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RATIONAL DESIGN OF TUNNEL SUPPORTS: AN INTERACTIVE GRAPHICS BASED ANALYSIS OF THE SUPPORT REQUIREMENTS OF **EXCAVATIONS IN JOINTED ROCK MASSES**

by

Michael D. Voegele

Department of Civil and Mineral Engineering University of Minnesota Minneapolis, Minn. 55455

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The Distinct Element methods portray a rock mass as a two-dimensional assembly of discrete blocks. There are no restrictions of block shapes or magnitudes of displacements and rotations. In the configurations used in this report, the Distinct Element method is coupled to a graphics terminal so that movements of the blocks are visually available as the computer calculates them. In Chapter II, a brief survey of the methods commonly used to analyze the behavior of jointed media is presented. Common to these methods surveyed is (Continued)

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the realization that the observed behavior of a jointed mass is different from the behavior of a continuum.

Chapter III is devoted to providing numerical verification of the Distinct Element method. In particular, several comparisons to limit equilibrium solutions are presented. The comparisons are favorable.

The other chapters are concerned with the behavior of a jointed rock mass when disturbed by an excavation. The discussion covers two broad topics: (a) excavations that are stable without external support, and (b) excavations that require external support. The behavior of the jointed mass is typically illustrated by means of contact force distributions within the mass and through the development of arching. For those excavations requiring support, computergenerated ground reaction curves are presented.

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This is the final report of a study performed by the University of Minnesota, Minneapolis, Minnesota, under Contract No. DACW45-74-C-0066 with the U. S. Army Engineer Waterways Experiment Station (WES), Vicksburg, Mississippi. This work was sponsored by the Office, Chief of Engineers, U. S. Army. This study, which was originally funded under the Civil Works Investigation Study (CWIS) Program, "Materials-Structures," by the Missouri River Division, Corps of Engineers, resulted in a report entitled "Rational Design of Tunnel Supports: A Computer Model for Rock Mass Behavior Using Interactive Graphics for the Input and Output of Geometrical Data." Following this preliminary study with its emphasis on rock mass behavior, the WES continued the contract under the CWIS Program, "Materials-Rock."

The study was conducted by Dr. M. D. Voegele, Department of Civil and Mineral Engineering, University of Minnesota, under the supervision of Professor Charles Fairhurst, Department Chairman. Technical contract monitor for the WES was Mr. J. B. Palmerton, Research Civil Engineer, Engineering Geology and Rock Mechanics Division (EG&RMD), WES. Dr. D. C. Banks, Chief, EG&RMD, was the Contracting Officer's Representative.

During the period of this contract and preparation of the report, the Directors of the WES were COL J. L. Cannon, CE, and COL N. P. Conover, CE. Technical Director was Mr. F. R. Brown.

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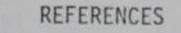
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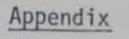
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CHAPTER I

INTRODUCTION

The goal of engineering analysis is intelligent design. This is true for disciplines which are based upon theoretical concepts discovered literally centuries ago as well as for more recently recognized disciplines such as Rock Mechanics engineering. Whereas the researcher in most fields of engineering has at his disposal analytical techniques which have been proven through decades of use and sound analytical development, the Rock Mechanics researcher has a limited number of analytical techniques at his disposal. Many of the problems encountered in the field of Engineering Geology and Mining engineering require the specification of the response behavior characteristics of a jointed rock mass. Foundation design requires a knowledge of the stiffness of the rock mass so that settlements and forces can be predicted accurately. Highway cuts in rock must be designed so as to be completely safe from slope failures. Mines, shafts and tunnels must all be designed with a knowledge of the behavior of the rock mass. The economic design of open pit mines relies heavily on the pit slope angle; a change of only a few degrees in the slope angle has a significant effect on the stripping ratio and thus the economic success of the mining venture. The design of dam foundations or abutments is particularly sensitive to

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the behavior of the rock mass. Settlements which can be tolerated by dam foundations are quite small. The failure to consider all of the response characteristics of a rock mass in such situations has in the past led to catastropic failures and the attendant loss of life. In all

of these problems the role of mass jointing can play a significant role

in the mass response, but all too frequently the exact behavior of the joints is poorly understood. Intelligent design requires an understanding of this behavior.

The analytic techniques at the disposal of the Rock Mechanics engineer upon which the design must be based are quite limited, and typically have been borrowed from other fields. The principles of classical mechanics are often used as an aid in analysis but it is frequently observed that the behavior of a rock mass cannot be characterized by the assumptions inherent in these classical methods. The fundamental assumptions of a continuum characterization, homogeneity and linearly elastic response, are often seen to be too limited in scope to characterize adequately the behavior of a rock mass. That group of materials which we classify as rock is typically non-homogeneous, anisotropic, and often discontinuous; of these characteristics the discontinuous nature of the rock mass is certainly the most influential in governing the ultimate behavior of the mass when subjected to some external stimulus. Constitutive relations can be generalized to include the effects of anisotropic structure; for example, a recent paper by Singh (1973) describes the development of an anisotropic continuum model in which the average influence of planar features can be taken into account.

Finite Element methods provide an accurate, approximate, method of solving problems in elasticity. The formulation of a "joint" element by Goodman et al. (1968) greatly increased the potential of the Finite Element methods in Rock Mechanics problems. However, Finite Element methods still strictly model a continuum and thus

large displacements are not possible except through iteration with each new iteration utilizing parameters derived from the previous iteration.

To portray adequately the response of a jointed rock mass requires the correct modeling of the discontinuities present, that is, the joints must have both normal and shear stiffness, they must obey some type of failure law and, most important, the blocks defined by the joints must be free to undergo large displacements and rotations if conditions so dictate. A computer model which satisfies all of these criteria was presented by Cundall (1971b).

The computer model for simulating progressive large scale movements in blocky rock systems which has since become known as the Distinct Element method utilizes semi-rigid rock blocks to characterize the behavior of a discontinuous rock mass. The interaction between the blocks is governed by realistic friction laws and simple stiffness parameters. There are no arbitrary limits on the amount of displacement and rotation allowed to each block and any block is permitted to touch any other block. True progressive failure is thus modeled and the mode of failure is automatically selected by the program since the system fails by that mode with the lowest stability. The program allows individual study of the effects of joint geometry, joint parameters,

loading conditions and excavation procedure.

The Distinct Element method portrays a rock mass as a two dimensional assemblage of discrete blocks. There are no restrictions on block shapes or magnitudes of displacements and rotations. In the configuration used in this dissertation, the program is interfaced with a graphics terminal so that movements of the blocks can be observed as the computer calculates them.

The equation governing the behavior of the blocks is solved in an explicit rather than implicit manner. Because the jointed rock mass may fail in such a way that the movement of the blocks leads to a new equilibrium position, an adequate block model must take this into consideration. An implicit solution assumes path independence; that is, the final answer must be the same no matter how the blocks move to get there. It seems safe to assume that path dependent phenomena such as separation along joints, stick-slip behavior of joint surfaces and block interlocking could not be modeled adequately except by an iterative procedure using very small time increments. It should be recognized that by using this approach, one would simply be using an implicit solution to model the solution that would have been obtained directly by an explicit approach.

The major approximation inherent in the Distinct Element method is that deformations occur along the surfaces of the rock blocks. This is accomplished by modeling each block as being rigid with what amounts to a thin elastic region around the perimeter. A consequence of this is that the program should produce the best solutions in situations where deformation is governed by movement along joint surfaces. On the other hand, those situations where elastic deformations of the rock mass are of the same order of magnitude as the movement along the joint surfaces are perhaps best modeled by elastic solutions of the Finite Element type or by a continuum characterization. Joint inclination and confining pressure play a significant role in the determination of the failure mode. The combination of the conditions of low confining pressures and favorable (or unfavorable dependent on viewpoint) joint orientation can lead to failure modes that are joint controlled. When viewed in terms of overall mass stiffness (i.e., deformation resulting from the application of external load), it can be seen intuitively that those failures in situations of low overall stiffness are probably joint controlled while the higher stiffness models exhibit failures that are essentially independent of jointing.

The research described in this dissertation has as its basis two main goals. First, owing to the relative newness of the Distinct Element method, a verification study has been undertaken to determine whether or not the Distinct Element method calculates solutions similar to other methods commonly used to analyze jointed rock masses. The second goal of the research is to apply the Distinct Element method to an engineering problem; in this particular case to the design of supports and the behavior of the rock mass surrounding an underground excavation. Underlying these two main research goals are several attendant yet equally important goals. One underlying theme concerns the application of computer interactive graphics to engineering analysis.

Another underlying theme concerns the potential perspective of the Distinct Element method. To introduce the investigations of the behavior of jointed rock masses performed with the Distinct Element method, a brief survey of the methods commonly used to analyze the behavior of jointed media is presented. Common to those methods surveyed is the realization that the observed behavior of a jointed mass is different than the behavior of a continuum. Several of the methods adopt the approach that the behavior of the jointed mass is fundamentally similar to that of a continuum; the same basic equations are assumed to govern both models but the constitutive relations are modified for the jointed models to simulate the presence of jointing. Other methods typically propound the fact that the jointing governs the mass behavior and thus postulate governing equations based upon assumed or observed behavior. This introductory section concludes with a brief overview of the Distinct Element formulation and presents several examples illustrating applications of the Distinct Element program.

Confidence in the use of approximate numerical techniques such as the Distinct Element method can best be developed by comparing calculated results to known solutions. However, for the particular case of the behavior of a jointed rock mass, comprehensive analytical solutions do not exist. The second major portion of this dissertation summarizes the results of numerous analyses, the sole purpose of which was to demonstrate the validity of solutions calculated by the Distinct Element method. The models chosen for comparison are typically simple and care was exercised to ensure that the behavior of the chosen model was described adequately by its solution. Most of the models chosen for the comparisons were based upon Limit Equilibrium principles, and the Distinct Element calculated solutions were seen to agree quite well with the Limit Equilibrium solutions in all cases. This general theme of comparison to existing solutions is not limited to this portion of the dissertation, however. Wherever possible in the later portions of the dissertation, every attempt is made to compare Distinct Element calculated solutions to other solutions.

The remainder of the dissertation is concerned with the behavior of a jointed mass when disturbed by an excavation. The discussion covers two broad topics: excavations which are stable without external support; and, excavations which depend upon externally applied support for stability. The interactive capabilities of the graphics terminal are fully utilized in these studies, both to observe the behavior of the mass and to modify the model while the program is running.

Chapter 4 presents the results of analysis of stable excavations in jointed rock. The behavior is illustrated by means of contact force distributions within the mass and interpreted as being governed by the development of arches within the mass. The mechanisms responsible for the development of the arching behavior are investigated and an interpretation utilizing arching theories is presented.

Chapter 5 presents the results of analyses of excavations in jointed rock which are not stable unless an external support is provided. The behavior is described quantitatively by ground reaction curves, relating the deflection of the excavation roof to the magnitude of the required support force. These curves reflect the interaction

between the rock mass and the support system in an attempt to guide the research along paths of investigation that are consistent with current thought regarding rational modeling of tunnel behavior. The results of these analyses are then compared to several methods, primarily of an observational nature, commonly used to design support systems for excavations in jointed rock. The rationale governing these comparisons is an attempt to provide some manner of analytic support for these routinely used design schemes.

The dissertation concludes with a summary of pertinent results and a critical assessment of the potential of the method in engineering analyses and design. The assessment of the potential emphasizes the limitation of the model in its present configuration with particular reference to the mini-computer based configuration. Suggestions for further development of the model are also presented, outlining areas of potentially fruitful research.

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CHAPTER II THE ANALYSIS OF THE BEHAVIOR OF A ROCK MASS CONTAINING PLANES OF DISCONTINUITY

2.1 Introduction

Before introducing the concepts underlying the Distinct Element model, a brief, historical review of the methods of analysis commonly used when dealing with the behavior of a discontinuous rock mass is presented. An exhaustive bibliography on jointed rock has been avoided, since a significant portion of all publications dealing with Rock Mechanics would need to be included. Rather, this chapter presents an overview of the methods of analysis used when dealing with jointed rock, concentrating on those methods that are accepted by engineers involved in actual design. The overview is relatively complete, including examples of all methods recognized to be in use at the present time.

A general survey of the response characteristics of a jointed rock mass is presented first, to enumerate those behavior mechanisms which must be incorporated in any analysis of a jointed rock mass if it is to portray accurately the behavior of the mass.

An overview of the methods of analysis is then presented. The

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methods lend themselves nicely to categorization in the following

groups:

Direct application of the principles of Soil Mechanics to the behavior of rock masses;

2) application of elastic theory, both in the classical

sense and by use of Finite Elements;

- behavior models including direct physical modeling as well as models based on observed behavior; and,
- methods of analysis utilizing Limit Equilibrium theories as developed in the fields of plasticity and soil mechanics.

The chapter concludes with a brief introduction to the Distinct Element method of calculating the behavior of a mass separated into distinct blocks by jointing or other discontinuity surfaces. The applicability of the model is discussed by way of a short presentation of worked examples. It is hoped that the examples selected give some insight into the scope and power of the method as well as demonstrating typical problems which can be analyzed by the method.

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2.2 The Response Characteristics of a Rock Mass

The obvious trend in the past several decades has been to excavations, both in mining ventures and the construction of civil works projects, on a scale never before attempted. The mining of vein type deposits frequently takes place in poor quality rock; in the case of the civil works projects, the best sites in terms of rock quality have already been selected for previous construction. Since it was no longer possible to ignore the rock behavior, the traditional concept of the soundness and stability of a rock mass had to be re-evaluated. In recognition of this requirement, a study group, the International Study Group for Geomechanics, was founded in Salzburg, Austria in 1951. The goal of this study group was to develop relations among all workers dealing with construction in rock and to develop a practical approach to the mechanics of rock masses.

The findings of the study group, which was succeeded by the International Society of Rock Mechanics in 1962, were presented by John (1962), and the following few paragraphs, quoted directly from John's paper, attempt to summarize the philosophy of the Salzburg group.

"Because the particular properties of rock as foundation and construction material deviate, in many respects, from those of other foundation materials, rock mechanics is compelled to follow its own course. The continuity of soil masses ... resulted in methods for analyzing a continuum, thus defining the concept of soil mechanics. In situ rock, however, contrary to the wide spread assumption in foundation engineering, is rarely homogeneous; rarely without mechanical discontinuities. Therefore, rock mechanics is, in most cases, to be a study of a jointed structure, of a discontinuum."

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The philosophy of the Salzburg group emphasizes the collaboration between civil and mineral engineers and geologists. The interrelation of engineers and geologists is readily apparent in the fundamental concepts of Rock Mechanics as outlined by John:

- "For most engineering problems, the technical properties of a rock mass depend far more on the system of geological separations within the mass than on the strength of the rock material itself. Therefore, rock mechanics is to be a mechanics of a discontinuum, that is, a jointed medium"
- 2) "The strength of a rock mass is considered to be a residual strength that, together with its anisotropy, is governed by the interlocking bond of the unit rock blocks representing the rock mass"
- 3) "The deformability of a rock mass and its anisotropy result predominately from the internal displacements of the unit blocks within the structure of a rock mass."

C. Jaeger (1964) presented a similar philosophy to that of John and noted that engineering calculations should take a far more detailed view of the actual state of the rock mass. Recognizing the inadequacy of the (then) present state of the art, he outlined a program of suggested research, emphasizing model tests and investigations of stress distributions in jointed media.

Fairhurst (1967), in assessing the influence of defects and discontinuities on the behavior of a rock mass noted that failure in a rock mass always begins at some structural defect and that the analysis of the behavior of the mass must consider: the

orientation and distribution as well as the magnitude of the applied forces; the distribution and orientation of structural defects with respect to the applied forces; and the energy available to cause continuing movement in the mass.

One final requirement of any method used to calculate the response of a jointed mass is that it should incorporate all of the kinematically possible failure modes. In addition to sliding on discontinuity planes, rotation of individual blocks about their centroids is also kinematically possible as reported in field exposures by Muller (1964) and DeFreitas and Watters (1973) and on a laboratory scale by Hoffman (1970). An analysis incorporating only force equilibrium and ignoring moment equilibrium could easily result in the neglect of an important response of the mass.

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2.3 Direct Application of Soil Mechanics Theories

Recognizing that large displacements preclude the use of elastic theory, Seldenrath (1951) idealized the strata comprising European coal measures as masses of loose structure, and attempted to apply Soil Mechanics principles to the problems of calculating fracture planes due to subsidence and calculating loads on props at a working longwall face. To the extent that he assumed reasonable values for friction coefficients, he was able to generate results that were confirmed in practice.

Morrison and Coates (1955) presented a method for the estimation of stresses surrounding a circular vertical shaft by means of plastic flow relationships deduced from Mohr's circle of stress. They questioned the utility of their method for practical design and concluded that although the approach was better than a simple elastic analysis, the actual material behavior was still more complex.

Wilson (1959) applied general Soil Mechanics principles to the problem of slope stability in open pit mines. He concluded that failures of cut slopes in fractured and fissured rock were often the result of uplift pressures in the water behind the slope face. Observing that the strength of granular material appeared to be

independent of particle size provided that a constant degree of compactness was maintained, Wilson extrapolated this result to the analysis of the behavior of broken and fissured rock. Since the scale of the jointing relative to the size of the pit was small, Wilson analyzed the stability of cut slopes using the principles of Soil Mechanics.

Jaeger (1970) analyzed highly jointed and broken rock by regarding the jointing as random and applying the laws of Soil Mechanics to its behavior. His analysis suggested that values of Youngs' modulus measured by plate bearing tests on jointed material for which the plate covered several joints were in reasonable agreement with laboratory values measured on actual specimens of the material containing many joints.

2.4 Elastic Theories Applied to Rock Masses

Elastic analyses of discontinuous or jointed masses can be conveniently grouped into two classes although the difference between the methods is one of application rather than fundamental difference in the theory. The first class comprises methods of analysis which directly utilize classical elastic theory; frequently the input parameters are modified to reflect different behavior modes due to the presence of discontinuities. The second class comprises Finite Element type analyses wherein the continuum is discretized and a stiffness relationship is formulated for applied forces and nodal point displacements. This latter class is obviously well suited to the situation of varying material properties throughout the mass.

2.4.1 <u>Classical continuum elastic theories</u>

Obert, Duvall, and Merrill (1960) restricted their analysis of the design of underground openings to competent rock but included horizontally stratified rock provided that the bond between layers was weak.

Beam and Plate theory were used for the analysis but it was noted that requirements of an elastically perfect, homogeneous, isotropic mass precluded the possibility of any fracturing in the roof unless it was parallel to the span direction. Barla (1970) presented constitutive relations for the nonlinear and time dependent behavior of rock masses but did not

present relations for discontinuous masses.

Smart (1970) developed a continuum model consisting of rigid cubical blocks set in a clay matrix and found good agreement with field data.

Singh (1973a, 1973b) used strain energy principles to derive general constitutive equations for a rock mass containing an arbitrarily oriented set of orthogonal, discontinuous joints in terms of a "stress concentration factor" matrix (which he computed by Finite Element analysis). His model gave good results for regions of low stress gradient but was found to give poorer results in regions of high stress gradient.

2.4.2 <u>Finite Element analyses</u>

One particular type of elastic analysis has gained acceptance since its inception. The Finite Element analysis, particularly in light of the modifications described below, has become a routinely used tool in Rock Mechanics problems.

Zienkiewicz et al. (1968) noted that linear elastic solutions indicating regions of tension in a rock mass were probably unrealistic for the general case of a cracked and fissured mass. Using a Finite Element formulation with an included "stress transfer" iteration they were able to calculate a solution with

no tension present in the mass. They also demonstrated that the solution provided a lower bound to the load at failure. Goodman, Taylor, and Brekke (1968) succeeded in incorporating a zero thickness element with normal and shear stiffnesses within the Finite Element formulation. With this special "joint element" they modeled failure in tension and shear, rotation, arch development and collapse patterns in jointed rock.

Hoffman (1970) compared the results of model tests with the results of Finite Element analyses and found that the large deformations and geometric changes in the jointed mass were not compatible with the assumptions inherent in the Finite Element method.

St. John (1972) analyzed the behavior of rock slopes in open pit mines using Finite Element models incorporating joint behavior. He concluded that the technique provided acceptable results provided small displacement theory was relevant but stressed the need for field data to verify the constitutive laws used in the program.

Chappell (1974 a; 1974 b), and Burman, Trollope, and Philp (1975) related the behavior of a jointed medium to rigid body displacements of block centroids. The modified Finite Element formulation replaced the elastic blocks with rigid ones and connected the block centroids with "joint" elements capable of modeling the combined block and joint responses of stress versus strain and moment versus rotation. Appropriate moduli were obtained by physical experiments.

Wang and Sun (1970 a, b) and Wang, Sun, and Ropchan (1972) used Finite Element analyses to determine stresses in gravity loaded open pit slopes. These stresses were then incorporated in a Limit Equilibrium analysis to determine the safety factor of the slope with respect to sliding on a preselected failure plane. Manfredini, Martinetti, and Ribacchi (1975) used Finite Element analyses of slopes to demonstrate the inadequacy of Limit Equilibrium methods in design. One interesting, though not unexpected, conclusion from their study was that the intact properties of the rock mass played very little part in the behavior of the jointed medium.

2.5 Jointed Mass Behavior Models

The jointed mass behavior models have been arbitrarily separated into three groups. The first comprises true physical models including both those models where similitude requirements are met and those whose purpose is simply to demonstrate the kinematics of failure. The second group, photoelastic modeling, is a sub group of the first group but owing to the special type of information it yields, is considered separately. The third group comprises theories of behavior which are primarily based upon either empirical data and the results of model tests or postulated behavior mechanisms.

2.5.1 Physical models

Lang (1964) used physical models for assistance in understanding the behavior of underground power stations. The most significant result of this research was aid in visualizing deformation behavior of jointed media.

Krsmanovic and Milic (1964) undertook a comprehensive series of tests to determine pressure distribution in a discontinuum subjected to external loads. Their results demonstrated that the pressure distribution was most sensitive to the original state of stress of the mass.

Trollope (1966) examined the behavior of a trapezoidal opening in a jointed rock mass. His work indicated two zones above the opening: a triangular "suspended zone" above the opening and a stable region outside of the "suspended zone".

Goldstein et al. (1966) investigated the behavior of models of jointed slopes by using a centrifuge. The goal of their research was to investigate the different failure conditions of slopes cut in jointed rock.

Fumagalli (1968) outlined the general principles of mechanical similitude including the incorporation of discontinuity surfaces for the proper physical scale modeling of problems in rock.

Edwards (1968) constructed a model of an open pit slope with wooden blocks as an aid to the interpretation of deformation measurements obtained in the field. An important conclusion of his work was that even though the models were not truly scaled they reproduced the measured phenomena better than an elastic analysis.

Gaziev and Erlikman (1971) embedded strain gauges in plaster blocks and built models to examine pressure distributions in discontinuous masses. They concluded that the state of stress is characterized by two "streams" of stresses following the directions of the principal joint sets.

Erguvanli and Goodman (1972) stressed the importance of kinematic models to observe possible failure modes, as well as scale models which could more accurately predict true behavior patterns.

Goodman (1972) outlined the use of the base friction model to

observe the kinematic behavior of rock masses containing

discontinuities.

Barton (1974) examined the deformation of discontinuous models consisting of approximately 40,000 blocks. Cut slopes were

excavated in the model after consolidation. The outcome of the experiments was compared to Finite Element analyses and photoelastic studies reported in the literature at that time. In all cases the "reasonable" behavior as predicted by theory failed to materialize.

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2.5.2 Photoelastic models

Lang (1961) used photoelastic models to study the effects of the presence of joints in the roof of an underground opening. He also presented some guidelines for rock bolting based upon patterns of stress transfer observed in bolted photoelastic models.

Maury (1970) examined the distribution of stresses in horizontally stratified masses by means of photoelastic models. He noted that the observed behavior was fundamentally different from that predicted by continuum theory.

Brcic and Nesovic (1970) analyzed detailed two dimensional models of dam foundations by photoelastic models. Their results suggested that the presence of discontinuities was a most significant parameter in the definition of the foundation bearing capacity.

Ergun (1970) performed a photoelastic analysis of a biaxially loaded plate with orthogonal joints and noted that the stress distribution was affected by: voids in the joints, the ratio of applied pressure, the joint inclination, and the stress history. Chappell (1973) investigated the interactions of underground openings in jointed media photoelastically. His conclusion was that the mechanisms of slip, rotation, and interlock controlled the load distribution. Furthermore, he noted that the interaction between a number of openings tended to accentuate these mechanisms.

2.5.3 Observational models

The observation of the behavior of discontinuous masses as well as the behavior of laboratory models has led to several theories of behavior which for lack of a better name are herein termed observational models. These observational models attempt to predict behavior in light of stress disruption/or redistribution across planes of discontinuity such as joints, or, in the case of soils, grain contact. They often utilize the information gained from model experiments or collected from real situations and extract response patterns which are postulated to hold for a large class of problems.

Terzaghi (1946) carried out tests in railroad tunnels in the eastern Alps by inserting wooden blocks of known strength properties in timber sets. On the basis of the results of these tests, he postulated the expected loads on tunnel supports as a function of the degree of jointing of the rock mass under consideration.

Trollope (1957, 1961) developed an arching theory of force

distribution within granular masses by a statical equilibrium analysis of a mass consisting of systematically packed, smooth, rigid spheres. He applied this theory to block jointed models to deduce general design principles. The same approach was used by Trollope and Brown (1965) to develop general equations for the distribution of pressure in a discontinuous mass beneath a strip loaded foundation.

Hyashi (1966) formulated an approach to determine the distribution of stresses in a fissured foundation in terms of the combined Pascal distribution. The effects of cohesion and frictional resistance were incorporated by means of an iterative application of Bousinesq's equation. His model recognizes a transient depth below which slip no longer occurs along joint planes. In the absence of cohesion or frictional resistance his model reduces to that postulated by Froelich (1933) who idealized the contact stresses in stacked cylinders as an assemblage of tiered, simple beams.

Lane (1961) and Lutton (1970) presented empirical charts relating slope height to inclination. Their data indicated trends, but they recognized that adverse geologic structure could invalidate the use of the charts.

Abel (1966) constructed a statistical model for the estimation of support loads in a tunnel from measured steel set loads, geologic and construction factors. He noted that although the principles of analysis were general, every tunnel must be considered as a separate problem.

Ross-Brown (1973) collected data concerning the stability of cut slopes in open pit mines throughout North America. He concluded that stability problems were too complex to be summarized by statistical relationships and that each mine needed to be considered as a separate entity in light of the experience obtained

in other mines.

More recently, Wickham, Tiedemann, and Skinner (1972), Bieniawski (1973), and Barton, Lien, and Lunde (1974) have presented empirically derived rock mass classification schemes for predicting loads on tunnel supports. The classification schemes result from the statistical manipulation of data collected during construction in rock and consider parameters such as joint spacing, orientation, infilling, and the presence of water.

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2.6 Limit Equilibrium Analyses

The basic principles of Limit Equilibrium applied to jointed rock masses are basically not different from the principles of the analysis of soil slopes as advocated by Fellenius (1936) or Bishop (1955). Owing to the degree of indeterminacy in the problem, assumptions must be made regarding the magnitude of some forces as well as their point of application.

A large portion of the literature on the stability of rock slopes comprises work on the analysis of the sliding behavior of tetrahedral wedges of rock by means of stereographic projection (e.g. John, 1968). Although two dimensional problems can be handled by this method, the amount of work required in the calculation as opposed to a simple graphical solution hardly merits the effort. Limit Equilibrium of three dimensional wedges is not considered in this review.

John (1962) presented a graphical analysis of the stability of a wedge of rock defined by joint planes and a cut surface. To determine the magnitude of rock anchor forces, he utilized conditions of limiting equilibrium by assuming that full frictional resistance would be developed along the plane of sliding effectively allowing him to specify the force polygon.

Bray (1966, 1967 a, b) substituted the equations for principle stress in the Mohr-Coulomb-Navier relation to develop the ratio of principle stresses at failure by sliding in a jointed mass as a function of the orientation of the principle stresses and the friction coefficient. An interesting outcome of this analysis comes by superposing a system of multiple fractures; in this model the value of the stress ratio approaches that of the active pressure coefficient as used in soil mechanics.

Jennings (1970) noted that failure in rock slopes did not necessarily follow a single plane. Rather, the failure surface that developed was often stepped. Utilizing Limit principles, the equations he presented incorporated sliding on a discontinuity as well as failure through intact rock.

Calder (1970) used Limit principles to analyze the stability of slopes in jointed rock. His analysis demonstrated that contrary to the case of slope failure in soils, significant changes in cut slope angle in jointed masses often have no effect on the degree of stability.

Hoek (1970) presented design charts, based on Limit Equilibrium principles, for the rapid assessment of the stability of slopes excavated in jointed rock. The assumptions necessary to produce the charts are conceded to be severe but are common to all analyses of this type.

Rosengren (1971) presented the results of a comprehensive analysis of the stability of blocks and wedges formed by the joint systems. Whereas the factor of safety as used by most investigators relates total driving force to total resisting force, Rosengren's definition of factor of safety contains one term relating available friction to required friction and another term relating required cohesion to available cohesion. Pentz (1971) investigated the situation where the failure criterion was not linear; a simple power law was used to relate normal stress to shear stress in place of the commonly used Mohr-Coulomb-Navier relationship.

Gaziev and Rechitski (1974) used Limit Equilibrium principles to analyze a rock slope with multiple slip modes possible. Their analysis located the layer with the minimum stability factor. The overall stability of the mass was then related to the individual layer stabilities.

Statistically based modifications of Limit Equilibrium methods have also been presented by several authors.

McMahon (1971) introduced design procedures that determine the probability that a rock slope will be undercut by joints that lie in unstable orientations. On the basis of these assumptions, and utilizing Limit Equilibrium principles, he arrived at curves relating probability of failure to slope angle.

Serrano and Castillo (1974) introduced probability density functions for the strength of discontinuities and the matrix as well as for block size and combined them with Limit Equilibrium principles to generate a stability curve for a rock slope in terms of probability of failure.

2.7 <u>An Evaluation of the Techniques Commonly used in</u> Jointed Mass Modeling

The preceding literature survey dealt with the numerous methods commonly used to predict the behavior of rock masses containing planes of weakness. It is of interest to present a brief summary of this survey that emphasizes what, in particular, advantages each of the methods offer.

The observational type methods are typically the first "analytical" method associated with engineering analyses. It is to the credit of men like Terzaghi that they recognized that the degree of jointing present in a rock mass could be the most significant factor to be considered in a design. However, most investigators pursuing this method noted that although the method usually worked quite well for a given problem, the information gained was generally not of use at other sites. Most recent investigators have tried to overcome this shortcoming by statistical manipulation of a large amount of data.

Elastic solutions, and in particular, modified elastic solutions are recognized as having shortcomings, but are usually conceded to be fairly accurate in those cases where the jointing is homogeneous throughout the rock mass. The modified solutions usually attempt to account for the jointing by anisotropic mass

behavior. It is interesting to note that one of the leading

proponents of this method of solution "... has now abandoned his

earlier view ... that an 'equivalent orthotropic medium' can be constructed to fairly represent the deformability of regularly jointed rock ..." (Goodman, 1974). Goodman makes this statement on the basis of dilatancy and stress dependent behavior of the joints and suggests that the more influential discontinuities should be treated as individual rock mass components.

The application of soil mechanics theories to the analysis of the behavior of jointed rock masses has been successful in those cases where the scale of the jointing relative to the problem was sufficiently small. However, if detailed analysis, on the scale of the jointing, is required, the method lacks validity.

The use of Limit Equilibrium principles holds much promise if it is possible to reduce the intricacies of the problem to the point where a "handleable" number of equilibrium equations can be written, and if the joint behavior may be represented as simply as is done in Limit Equilibrium methods. The main problem with this type of approach is that the necessary assumptions often tend to oversimplify the problem - if too many assumptions need to be made to reduce the indeterminacy, then the model may no longer be representative of the problem to be solved.

Physical modeling seems to offer the best solution to modeling the behavior of jointed rock masses, since the behavior is exactly modeled if similitude requirements are met. However, it is

virtually impossible to set up the identical physical models which are necessary for parametric variation, and the cost of a detailed model can be prohibitive. The Distinct Element method offers a combination of the

capabilities required to predict the behavior of jointed rock

masses. The joints are modeled as the most significant components of the problem. There is no need to oversimplify the problem and the data structures can be stored permitting a given geometry to be analyzed as many times as desired.

It is in the context of a reproducible "physical" model that the Distinct Element method is used in this dissertation.

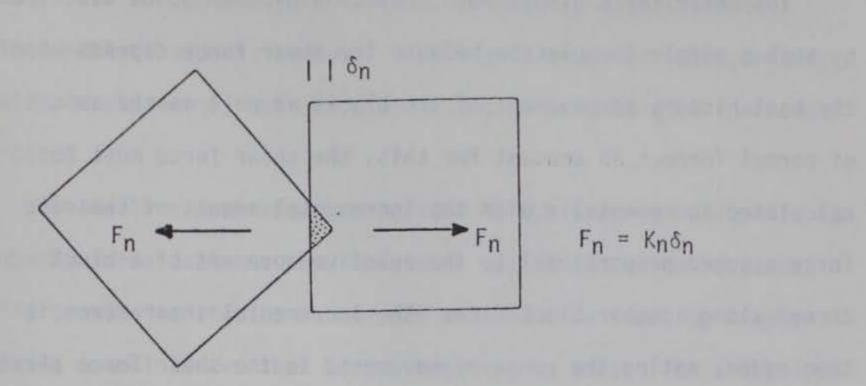
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2.8 The Distinct Element Method

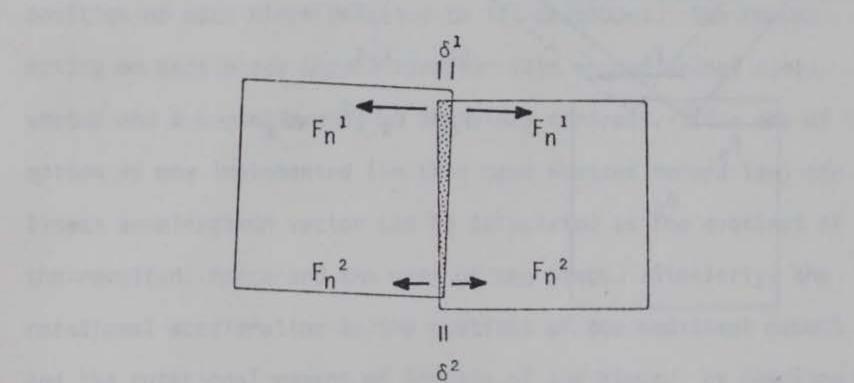
The Distinct Element method introduced by Cundall (1971 a, b) is a computer based analysis that simulates the behavior of a system of discrete, semi-rigid rock blocks. Block interactions are governed by realistic friction and stiffness laws. Each block may undergo unlimited displacement and rotation while progressive failure is modeled. In its present formulation the program is run in an interactive mode on a dedicated mini-computer coupled to a cathode ray tube (CRT) graphic output device. The CRT is used both for the input of geometric and material information as well as for the output data which consists of drawing the movements of the blocks as a function of time. The description presented follows Cundall (1971 b).

The program calculation cycle comprises force-displacement relations for the block contacts and laws of motion for the block centroids. Very simple relationships are used to relate normal force to normal displacement and shear force to shear displacement.

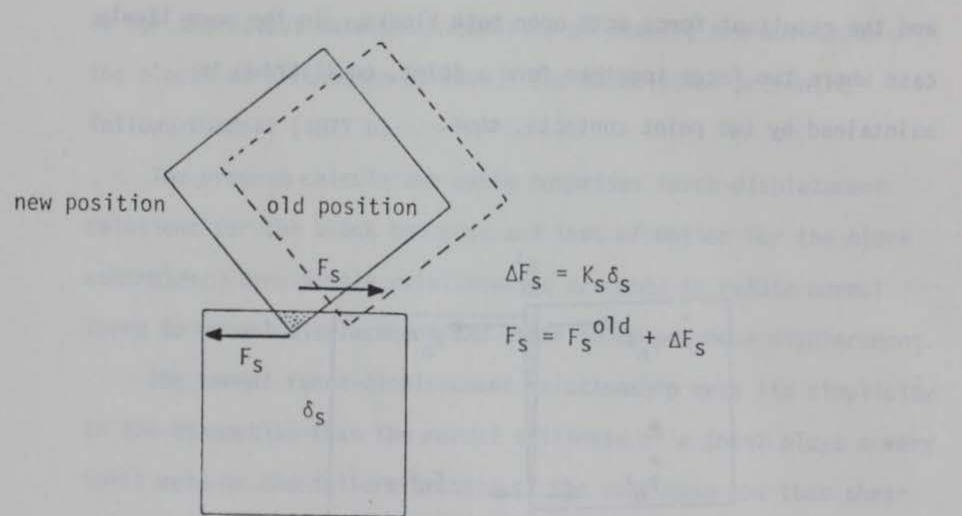
The normal force-displacement relationship owes its simplicity to the assumption that the normal stiffness of a joint plays a very small role in the failure process of the rock mass and that shear force does not affect normal force. Thus normal force is assumed proportional to the overlap between two blocks. Diagramatically,



where constant of proportionality K_n is the joint normal stiffness and the resultant force acts upon both blocks. In the more likely case where two faces together form a joint, equilibrium is maintained by two point contacts, thus:



Cundall argues for the validity of representing a joint by two point contacts by noting that owing to irregularities present on a real joint, contact will occur only at discrete points, quite possibly only two. The shear force-displacement relationship cannot be described by such a simple formulation because the shear force depends upon the past history of movement of the blocks as well as the amount of normal force. To account for this, the shear force must be calculated incrementally with the incremental amount of shearing force assumed proportional to the relative movement of a block corner along another block face. The incremental shear force is then added, noting the sense of movement, to the shear force already existing between the two blocks. Diagramatically:



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where the proportionality constant K_s is the joint shear stiffness. Although not strictly necessary from a physical standpoint, the normal force is also calculated incrementally in the program

so that all forces are derived from incremental displacements. This formulation does, however, simplify the task of incorporating nonlinear phenomena, such as dilatation, associated with the normal stress.

Two failure laws are incorporated in the program. Since it is probably unrealistic to have tensional resistance across a joint, a "no tension" criterion is adopted at each time step, by simply setting normal forces that become negative to zero. The criterion governing shear failure is the Mohr-Coulomb-Navier law. At every time step, the shear force at each contact point is tested and limited to a maximum force, which is dependent upon the normal force.

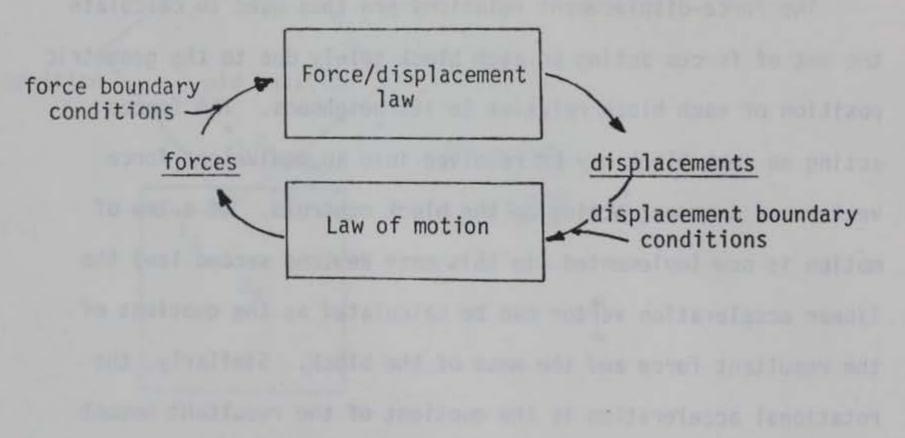
The force-displacement relations are thus used to calculate the set of forces acting on each block solely due to the geometric position of each block relative to its neighbors. The forces acting on each block may be resolved into an equivalent force vector and a moment acting on the block centroid. If a law of motion is now implemented (in this case Newtons second law) the linear acceleration vector can be calculated as the quotient of the resultant force and the mass of the block. Similarly, the rotational acceleration is the quotient of the resultant moment and the rotational moment of inertia of the block. By choosing a suitable time step, these accelerations may be numerically integrated twice to give the displacement of the block. For example, in the x direction:

$$v_{x}^{new} = v_{x}^{old} + \frac{F_{x}}{m} \cdot \Delta t$$

 $u_{x}^{new} = u_{x}^{old} + v_{x}^{new} \cdot \Delta t$
 $v = velocity$
 $u = displacement$
 $m = mass$
 $F_{y} = Force on block in x dir$

with similar equations for the y direction and rotation. The time step cannot be made arbitrarily large, or rapid geometric changes would not be modeled accurately. However, a more subtle reason for the limit on the time step is that owing to numerical instabilities in the solution of the equations, there is a limit to the maximum time step. This is discussed in more detail by Cundall (1971 a) along with the damping requirements of the equations.

The complete calculation cycle can be summarized as:



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In addition to the main calculation cycle, routines are needed to keep track of the coordinates of contacts; the use of arbitrarily large displacements and the attendant large number of possible contact points requires the implementation of a dynamic memory allocation scheme. This scheme is discussed in Appendix B along with a more complete listing of the equations comprising the main calculation cycle. A complete discussion of the fundamental algorithm of the program is given by Cundall (1974).

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2.9 Applications of the Distinct Element Method

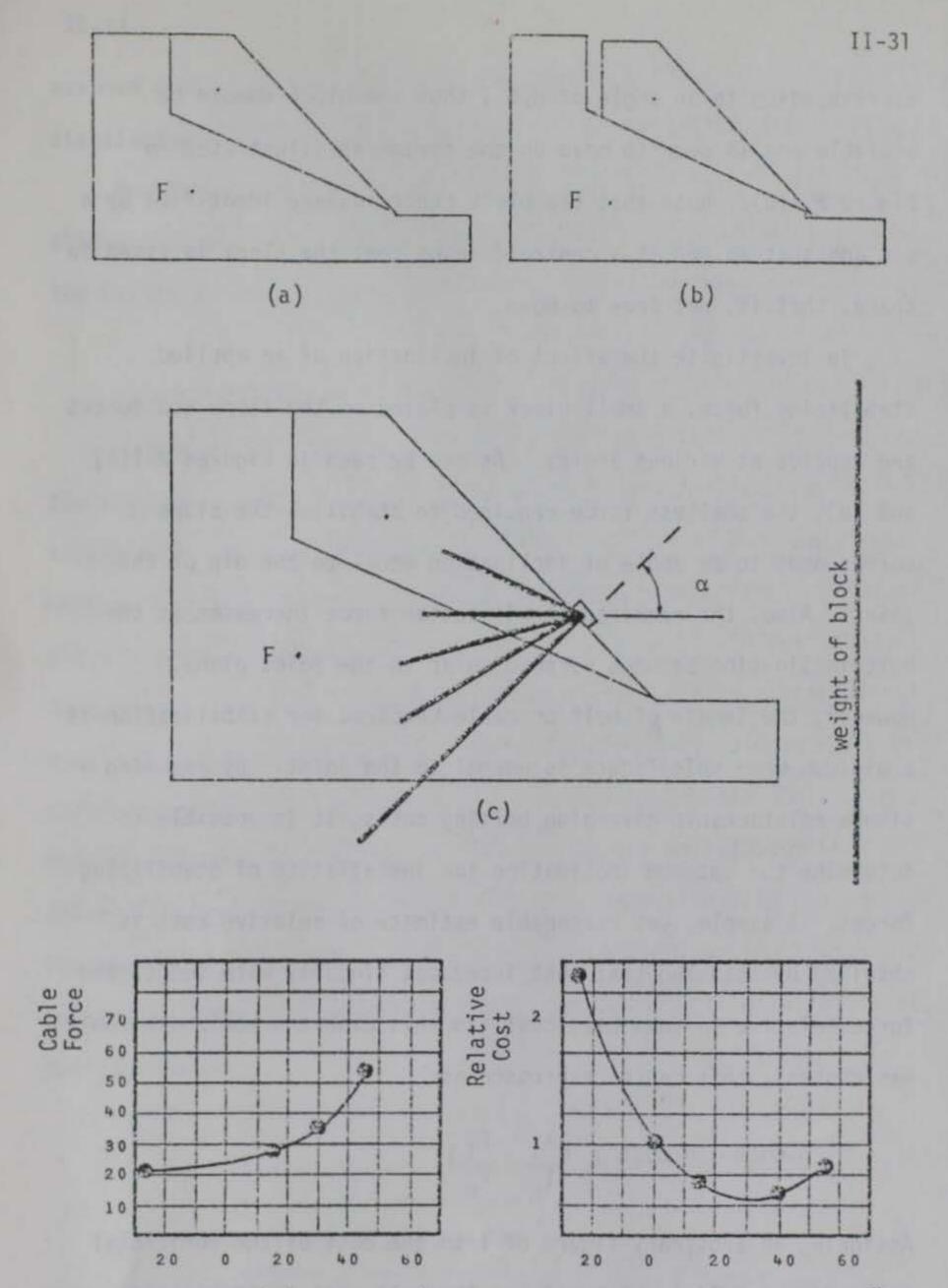
As a conclusion to this chapter, several examples illustrating the application of the Distinct Element method to problems involving the response behavior of jointed rock masses are presented. The problems range in complexity from modeling a rock slope as a single block bounded by a joint plane and a tension crack at the crest, to examining the behavior, as failure progresses, of a jointed mass being mined by caving techniques. The examples chosen illustrate most of the salient features and capabilities of the Distinct Element method; however, the potential of the method extends much farther. Particular examples of extended applications could include true blasting analysis, coupled fluid flow behavior and incorporation of elastic stresses and strains.

The problem of the correctness of the solutions obtained by the Distinct Element method will be addressed in the next chapter; for the present time the correctness of the solutions should be accepted. Alternatively, the examples can be viewed in light of kinematics only with calculated displacement modes and forces interpreted in light of experience and intuition.

Example 1 - Stabilization of a Failing Rock Slope

The rock slope illustrated in Figure 2.1(a) consists of a

single block bounded by a joint plane dipping approximately 25° out of the face of the slope and a vertical tension crack at the crest of the slope. The friction coefficient of the joint plane is .15,



(d) ^α (e) ^α

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Figure 2.1 Stabilization of a Failing Rock Slope

corresponding to an angle of 8.5°; thus the block should be unstable and is seen to move on the screen as illustrated in Figure 2.1(b). Note that the block centroids are identified by a dot and that an "F" at a centroid means that the block is fixed in space, that is, not free to move.

To investigate the affect of inclination of an applied stabilizing force, a small block is placed on the slope and forces are applied at various angles. As can be seen in Figures 2.1(c) and (d), the smallest force required to stabilize the slope corresponds to an angle of inclination equal to the dip of the joint. Also, the required stabilization force increases as the bolt inclination becomes perpendicular to the joint plane. However, the length of bolt or cable required for stabilization is a minimum when this length is normal to the joint. By assuming a simple relationship governing bolting costs, it is possible to determine the optimum inclination for installation of stabilizing forces. A simple, yet reasonable estimate of relative cost is obtained by assuming that cost increases linearly with length and force relative to some base cost (in this case the horizontal bolt was chosen), this can be expressed as:

Cost i = Cost H
$$\left(\frac{1}{1} \cdot \frac{1}{F}\right)$$

Assigning an arbitrary figure of 1 to the cost of the horizontal bolt, Figure 2.1(e) which relates the bolt cost to inclination, can be plotted. From this figure it can be seen that based upon the

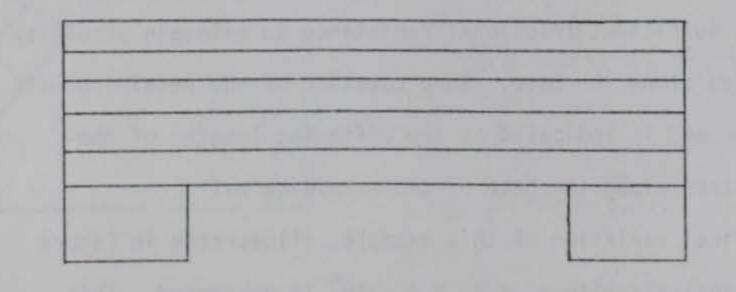
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assumed cost relationship, the optimum angle of inclination of the stabilizing force is approximately 30°.

Realistic cost data can be used to refine the cost relationship and much more complicated slope geometries can be modeled with the Distinct Element method.

Example 2 - Horizontally Stratified Mine Roof

Figure 2.2 illustrates a horizontally stratified mine roof; there are no joints exposed within the span of the roof. The only information that can be obtained by using the Distinct Element method in a problem such as this is the weight distribution on the pillars which in this case could readily have been obtained by inspection. The Distinct Element method in its present formulation does not incorporate elastic behavior of the elements; all deformations occur on joint surfaces. For problems where elastic deformations are important an elastic analysis such as Finite Element analysis should be used. For this particular problem however, beam theory could have been used to determine the bending moments and deflections (see, for example, Obert, Duvall, and Merrill 1960).



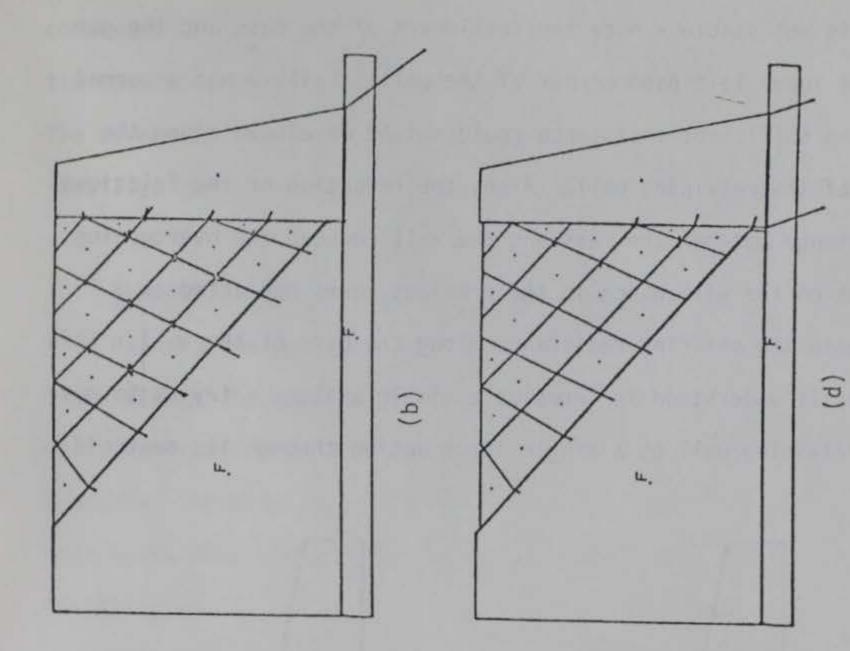
* Figure 2.2 A Horizontally Stratified Rock Mass

Example 3 - A Gravity Retaining Wall

Illustrated in Figure 2.3(a) is a retaining structure which is required to prevent movement of the jointed mass to its left. Three friction coefficients are involved in a problem such as this: ϕ , the friction angle of the joints within the mass; ϕ_b , the friction angle for sliding on the base of the wall; and, ϕ_w , the friction angle for sliding of the rock mass along the wall. By selectively varying these parameters it is possible to illustrate several aspects of the behavior of the wall in response to loading. Figure 2.3(b) illustrates the behavior of the wall when $\phi = 26^{\circ}$ and $\phi_b = \phi_w = 45^{\circ}$; as the blocks begin to move outward, the wall cannot slide along its base and thus begins to rotate as evidenced by the single contact vector at the lower right hand corner of the wall. The lower left hand corner of the retaining wall is actually lifted off the plane of sliding. The situation is, however, stable.

In Figure 2.3(c) another stable situation is illustrated. In this case, $\phi = \phi_b = 19^\circ$ while $\phi_w = 45^\circ$. The "9" printed on a surface indicates that that surface is assigned the friction behavior specified for material type 9. This analysis indicated that as the rock mass moved outward the base of the retaining wall moved until sufficient frictional resistance to maintain stability was generated along the base. Some rotation of the retaining wall

has occurred and is indicated by the differing lengths of the contact vectors along the base of the retaining wall. As a final variation of this example, illustrated in Figure 2.3(d), an analysis with $\phi_w = \phi_b = \phi = 19^\circ$ is presented. This



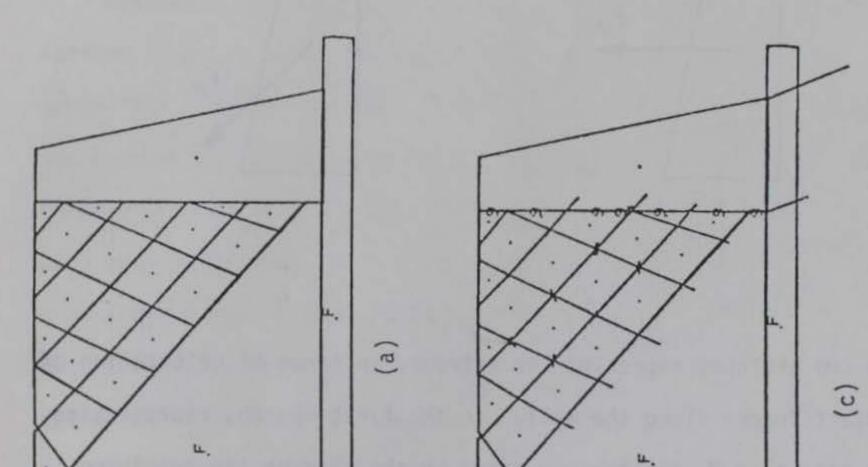
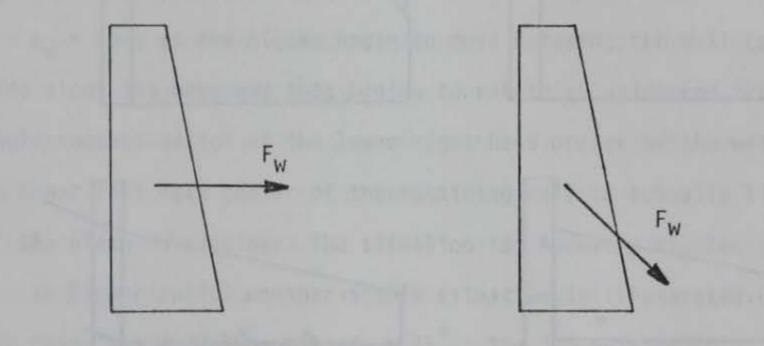


Figure 2.3 A gravity retaining wall

case is not stable - note the settlement of the mass and the gap at the lower left hand corner of the wall. Failure has occurred because sufficient resistance could not be developed along the base of the retaining wall. Also, the reduction of the frictional resistance between the mass and the wall reduced the overturning moment on the wall which in the previous cases had acted to increase the shearing resistance along the base of the wall. This is easily understood in terms of a simple analogy - trying to move the retaining wall by a single force acting through its centroid.



The two sketches represent the extremes in terms of orientation of contact forces along the wall. In the first sketch, representing the case $\phi_w = 0$, the force exerted by the mass on the retaining

wall, F_w , has no vertical component while in the second sketch, representing the case $\phi_w = 45^\circ$, the force exerted by the mass on the retaining wall, F_w , has a vertical component. The vertical

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component of F_W acts to increase the normal force on the base of the retaining wall, thus increasing resistance to sliding movement. The effect of increasing the coefficient of friction ϕ_W is thus to stabilize the retaining wall against translational sliding.

Example 4 - A Rock Slope Which Fails by Toppling

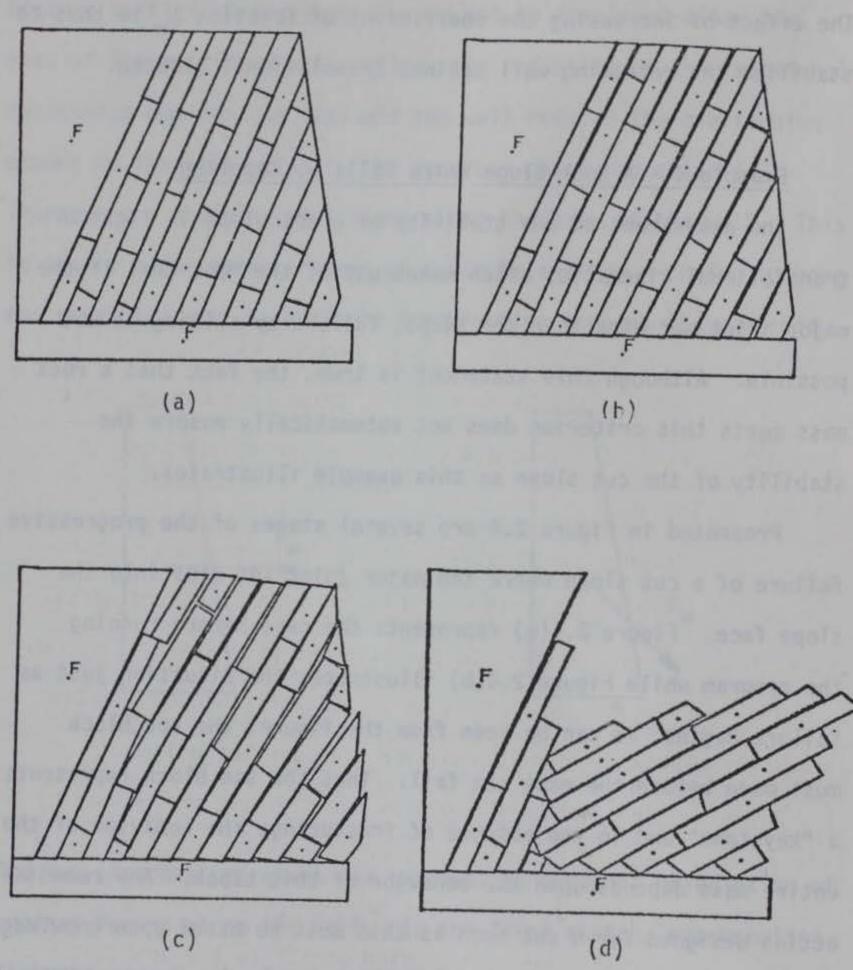
The assessment of the stability of a cut slope in light of translational kinematics often makes use of the fact that if the major joint set dips into the slope, failure by sliding is not possible. Although this statement is true, the fact that a rock mass meets this criterion does not automatically ensure the stability of the cut slope as this example illustrates.

Presented in Figure 2.4 are several stages of the progressive failure of a cut slope where the major joint set dips into the slope face. Figure 2.4(a) represents the case before running the program while Figure 2.4(b) illustrates the situation just as failure begins; as can be seen from the figure, the toe block must move before the mass can fail. Thus the toe block represents a "keystone" and in the absence of fracturing, the behavior of the entire mass depends upon the behavior of this block. Any remedial action designed for a cut such as this must be based upon knowledge

of which blocks or sections of the slope act as keystones. With the Distinct Element method it is a simple matter to determine which blocks can best be utilized to stabilize the mass. Figure 2.4(d) illustrates another physically observed feature which is accurately modeled by the Distinct Element method. After



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Figure 2.4 A rock slope which fails by toppling

a significant amount of movement has occurred, stable equilibrium of the mass is reached. (Blocks which moved away from the mass were erased as the program progressed).

Example 5 - Anchoring a Large Force in Rock Mass

This example presents a comparison of the failure loads calculated when a large external force, such as an anchorage force for a transmission tower, is applied to a jointed mass in two different directions. The rock mass in question and the two loading directions are illustrated in Figures 2.5(a) and 2.5(c). The force vectors which cause failure, drawn to a common scale, are also illustrated; the deformed geometries are illustrated in Figures 2.5(b) and 2.5(d).

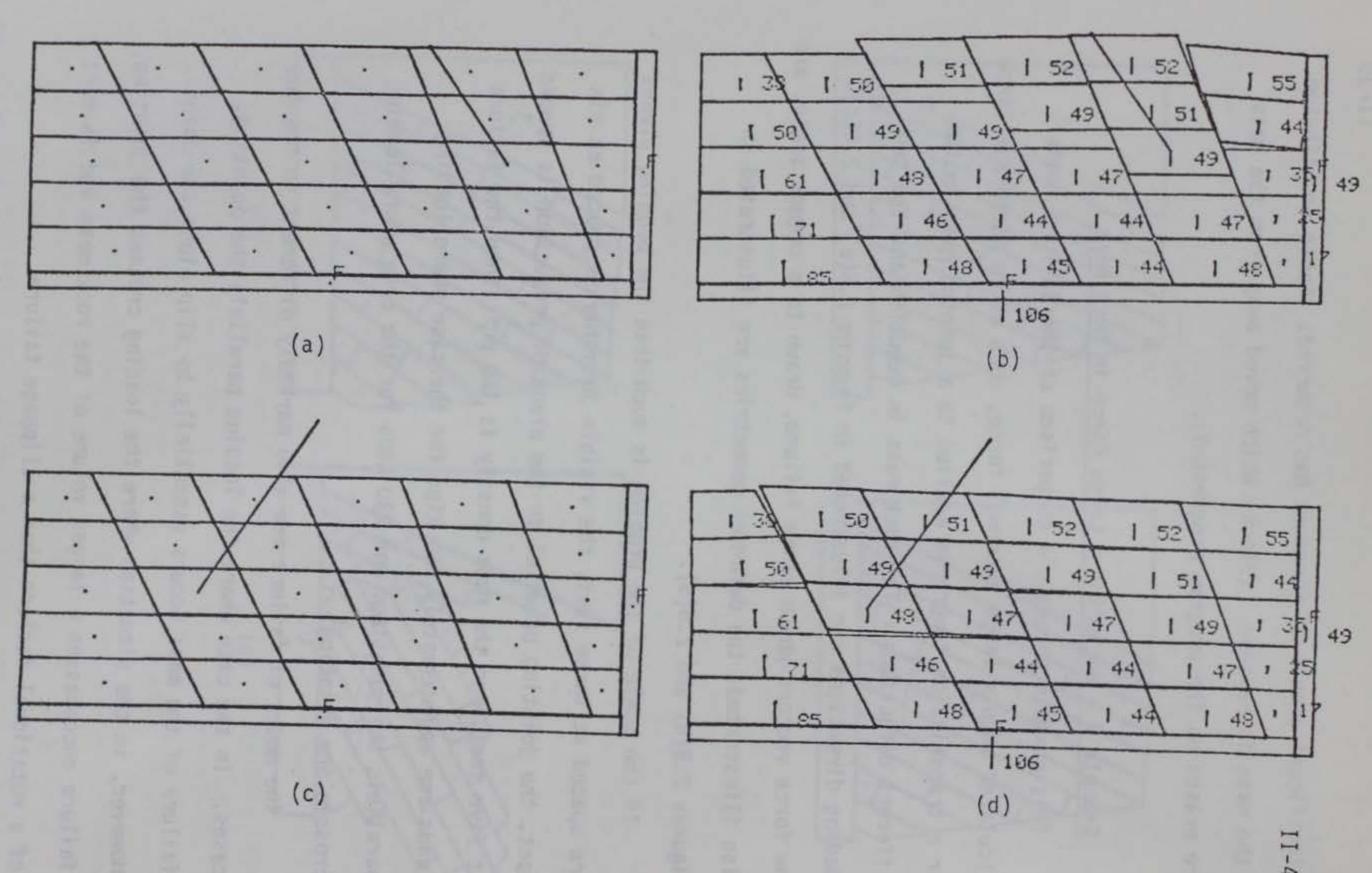
If the scale of the problem is such that the bedding planes are spaced at three feet, the visible jointing is spaced at six feet, the jointing parallel to the plane of projection is spaced at five feet, and the mass density is 160 pcf; then the failure loads are approximately 160 kips for the case where loading parallels the jointing, and 230 kips for the case where loading crosses the jointing.

The modes of failure are also markedly different in the two cases. In the case where the loading parallels the jointing,

failure of the mass occurs essentially by slip along the joints. However, in the situation where the loading crosses the jointing, failure encompasses a larger volume of the rock mass and is more of a rotational failure than a slippage failure.

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Figure 2.5 Anchoring a large force in a rock mass



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Example 6 - A Pressure Tunnel Near a Free Surface

This example examines a hypothetical situation where a pressure tunnel is located near a free surface. A situation such as this could be encountered, for example, in a diversion tunnel for a dam.

The failure of the rock mass in this particular case depends upon the penetration of water into the joints at fairly high pressures. Hopefully, in a real situation, water pressure testing would have been performed to assess the permeability of the mass and appropriate remedial action such as grouting and lining undertaken to prevent water loss. Nevertheless, the example is instructive and is presented in spite of its lack of realism.

Figure 2.6(a) illustrates the tunnel under consideration; the diameter of the tunnel is 20 feet and the internal pressure, which is assumed to penetrate all joints intersecting the tunnel, is 100 psi. The initial failure with the friction angle equal to 22 degrees on the joint planes is illustrated in Figure 2.6(b). In this type of problem the water pressure does not decrease as the joints open, for there is a practically unlimited supply of water to move out into the joints as they open.

Figure 2.6(c) shows a later stage of the progressive failure while Figure 2.6(d) illustrates the pressure distribution in the

joints as indicated by an asterisk on those joints where water

pressure is applied. The water pressure units illustrated are

internal computer units and are seen to follow a parabolic trend,

decreasing in intensity from the tunnel to the free surfaces. The

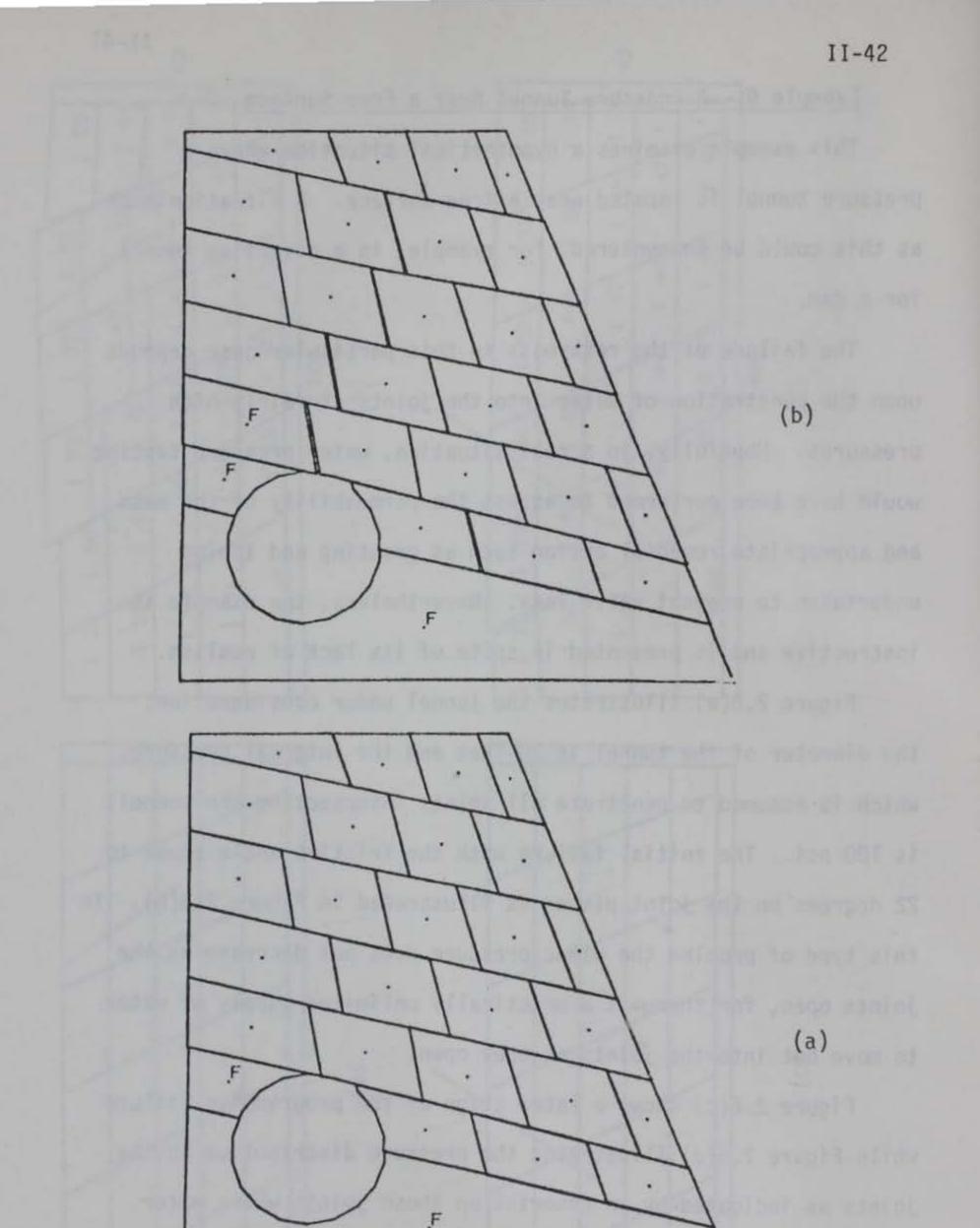
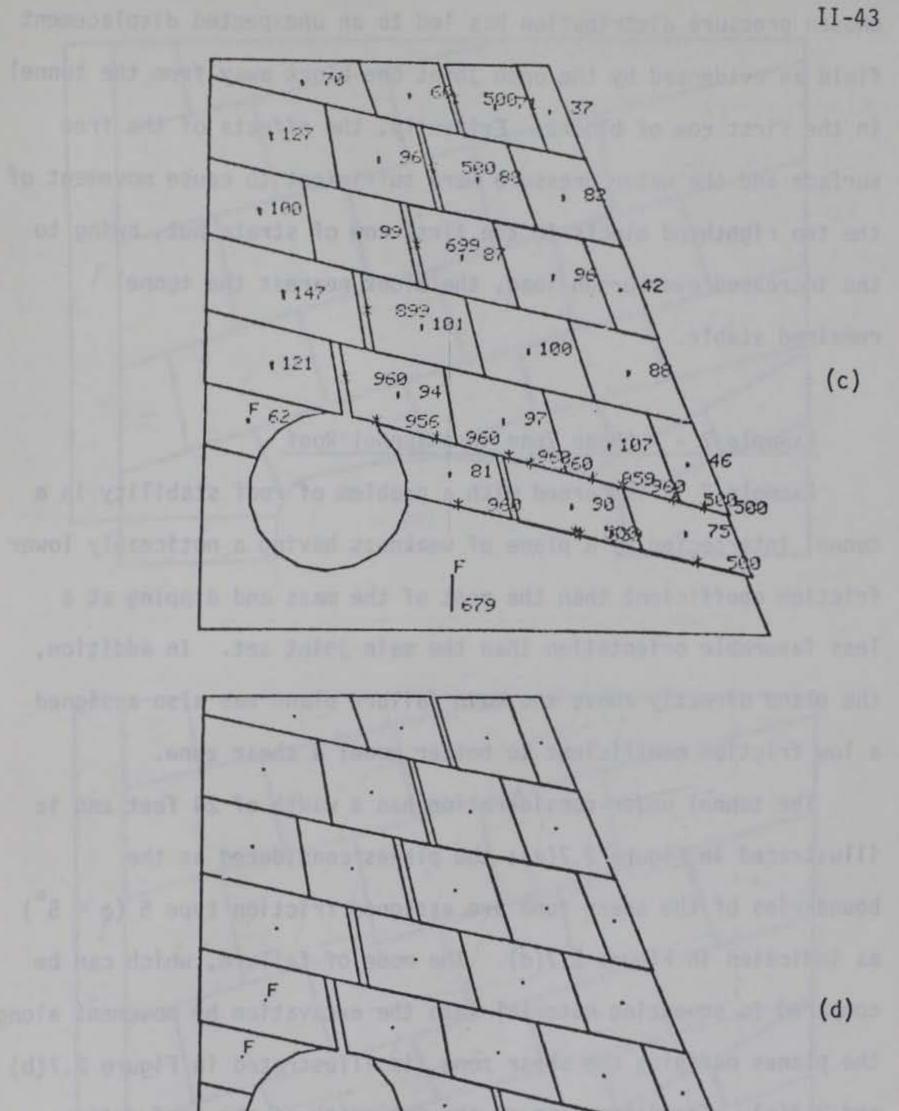
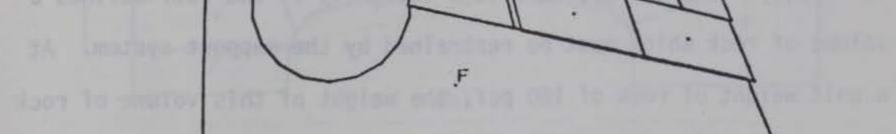


Figure 2.6 A pressure tunnel near a free surface



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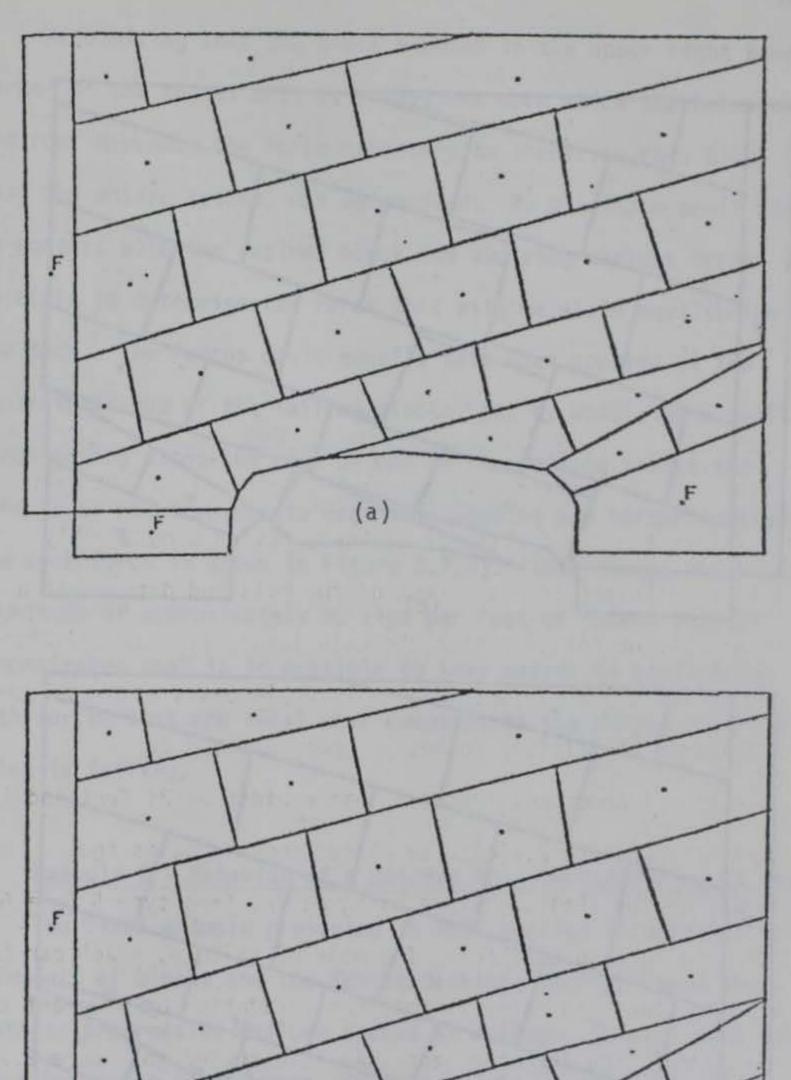
chosen pressure distribution has led to an unexpected displacement field as evidenced by the open joint one block away from the tunnel in the first row of blocks. Evidently, the effects of the free surface and the water pressure were sufficient to cause movement of the two righthand blocks in the first row of strata but, owing to the increased overburden load, the block nearest the tunnel remained stable.

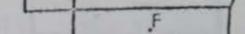
Example 7 - A Shear Zone in a Tunnel Roof

Example 7 is concerned with a problem of roof stability in a tunnel intersected by a plane of weakness having a noticeably lower friction coefficient than the rest of the mass and dipping at a less favorable orientation than the main joint set. In addition, the plane directly above the main failure plane was also assigned a low friction coefficient to better model a shear zone.

The tunnel under consideration has a width of 24 feet and is illustrated in Figure 2.7(a); the planes considered as the boundaries of the shear zone are assigned friction type 5 ($\phi = 5^{\circ}$) as indicated in Figure 2.7(d). The mode of failure, which can be compared to squeezing material into the excavation by movement along the planes defining the shear zone, is illustrated in Figure 2.7(b) and 2.7(c). The disruption of the integrity of the roof defines a

volume of rock which must be restrained by the support system. At a unit weight of rock of 160 pcf, the weight of this volume of rock is approximately 100 kips per foot of tunnel length.





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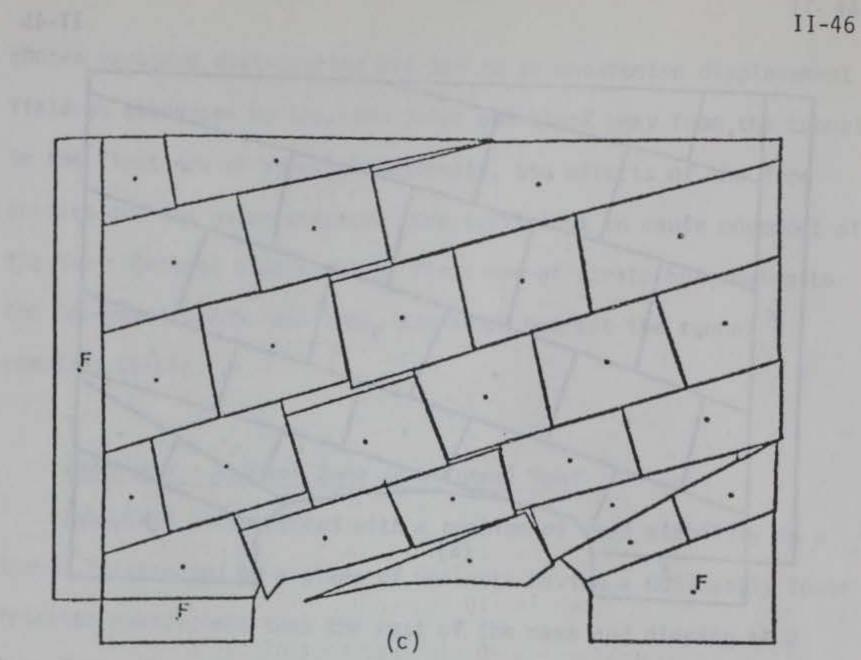
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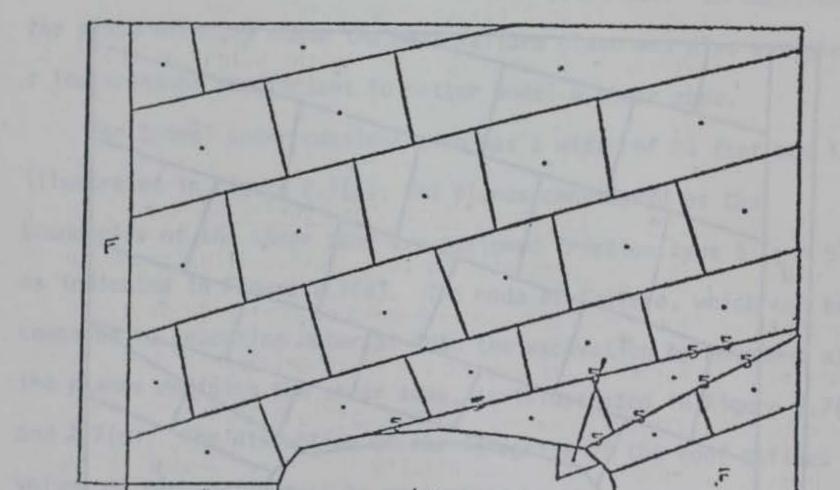
Figure 2.7 A shear zone in a tunnel roof

tention of the second and their addition for the same of the second in the

(b)







÷

The of the per during and all the state of the Figure 2.7 Continued

(d)

Recognizing that the block exposed in the upper right hand corner of the tunnel acts as a keystone upon which the behavior of the roof depends, the force necessary to stabilize this block (and thus the entire system) was determined. By placing a small block in contact with the desired block and applying various forces it is possible to determine the force that will maintain equilibrium of the mass. The forces could equally have been applied at the centroid of one of the failing blocks, but by utilizing a small block acting along the edge of one of the failing blocks the effects of rotation due to eccentric loading are better modeled. One such force is shown in Figure 2.7(d). This force, which has a magnitude of approximately 20 kips per foot of tunnel length demonstrates that it is possible to keep masses in equilibrium with forces that are small when compared to the weight of the mass which is failing.

Example 8 - Behavior of a Jointed Mass During Mining by Caving

The final example presented in this section illustrates the movements of blocks and the forces developed during these movements as progressive failure occurs in a large, jointed mass being mined by caving techniques. The block configurations as mining

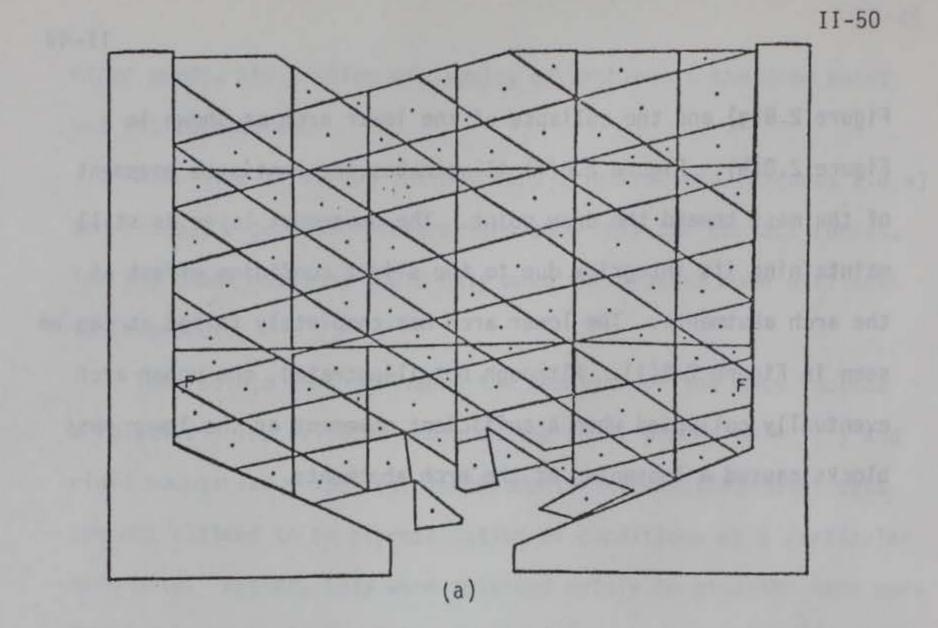
progresses are illustrated sequentially in Figures 2.8(a) through 2.8(j). The figures present the situation beginning some time after mining had commenced; in addition, as soon as individual blocks had moved sufficiently far from the mass so that they no longer influenced the behavior of the mass, they were erased. In other words, the problem of jamming or arching at the draw point was not considered.

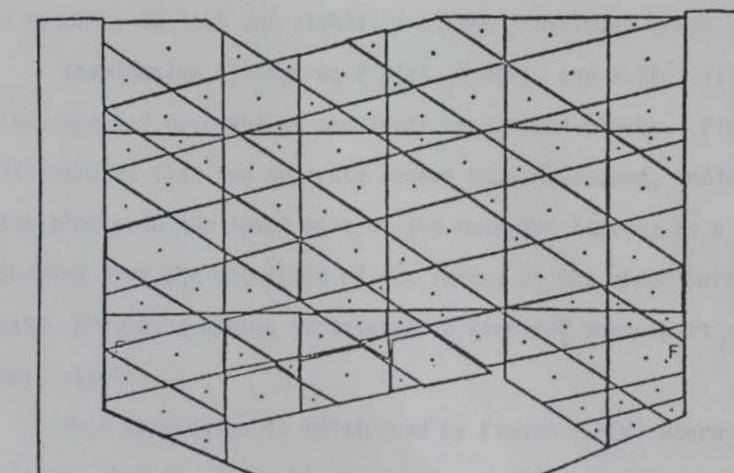
After the first two introductory illustrations (Figures 2.8(a) and 2.8(b)) alternate illustrations show only the contact forces, for the block outlines would only make the drawing more difficult to interpret.

The factors that influence the behavior of the mass include a relatively low friction angle on the joint planes ($\phi = 17^{\circ}$) and rigid boundaries. The four independent, intersecting joint sets are not claimed to be representative of conditions at a particular mine site. Rather, they were selected solely to give the mass more freedom to move, as two intersecting joint sets were found to have a tendency to lock and stabilize as the individual blocks moved.

Examination of Figures 2.8(a), 2.8(b), and 2.8(c) illustrate the expected movement of the lower unconfined blocks. Figure 2.8(d) illustrates that two separate arches have developed, indicating that the blocks in the lower part of the mass are failing as a unit and, judging from the magnitude of the forces in the upper part of the mass, providing enough resistance to keep the upper part of the mass stable.

This conclusion is reinforced by Figure 2.8(e) where it can be seen that the lower blocks are separating significantly from the mass. Figure 2.8(f) shows the continued development of two separate arches. The thrusts developed in the lower arch are not of sufficient magnitude to stabilize the mass, as evidenced by the progression of raveling up into the mass as illustrated in Figure 2.8(g) and the collapse of the lower arch as shown in Figure 2.8(h). Figure 2.8(i) illustrates the continued movement of the mass toward the draw point. The uppermost layer is still maintaining its integrity due to the slight confining effect at the arch abutments. The lower arch has completely failed as can be seen in Figure 2.8(j). Although not illustrated, the upper arch eventually collapsed when a sufficient movement of the lower mass blocks caused a loosening at the arch abutments.





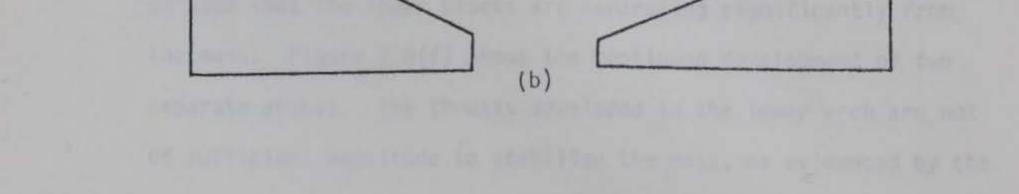
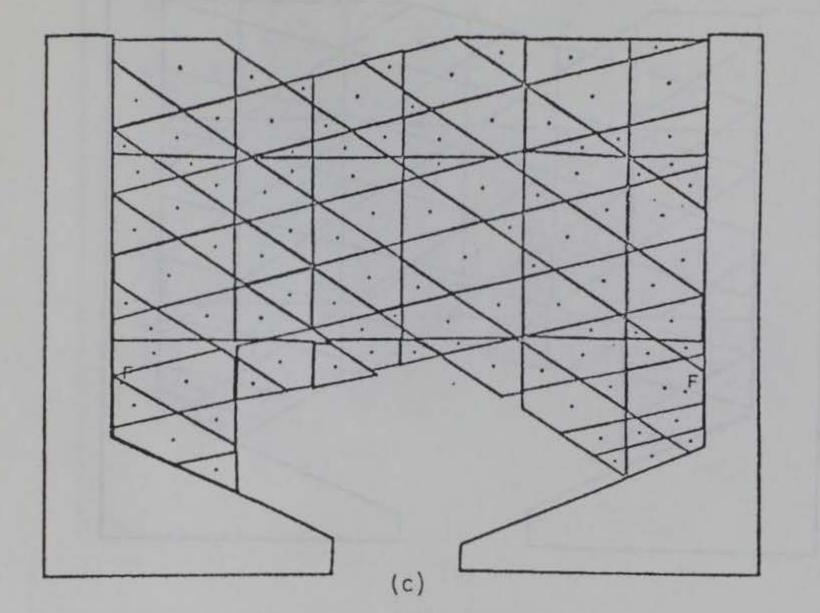
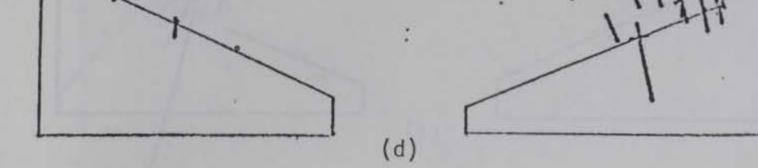
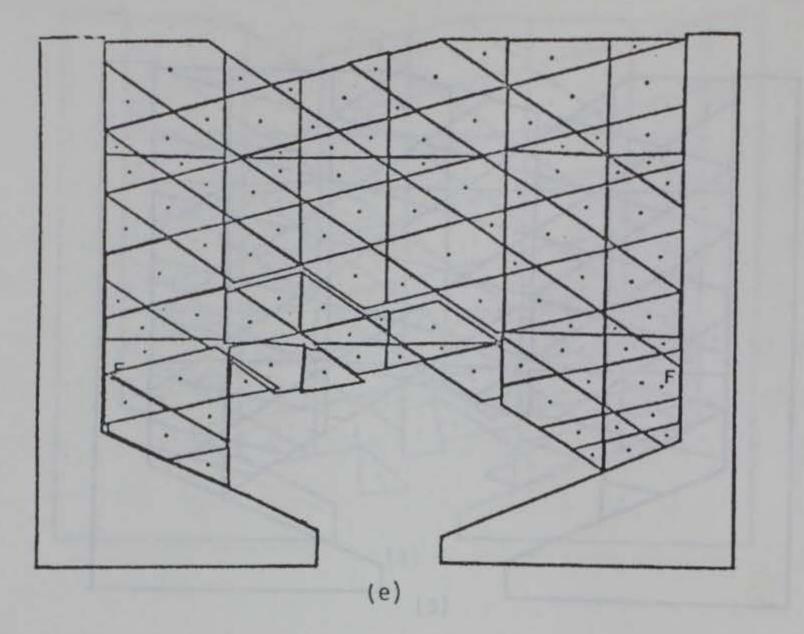


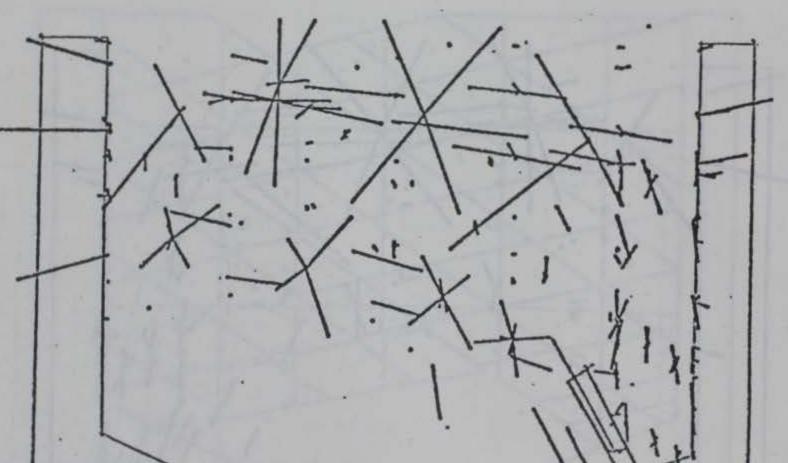
Figure 2.8 Behavior of a jointed mass during mining by caving

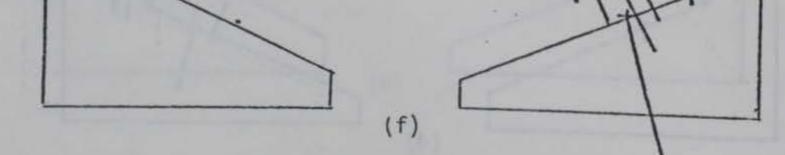


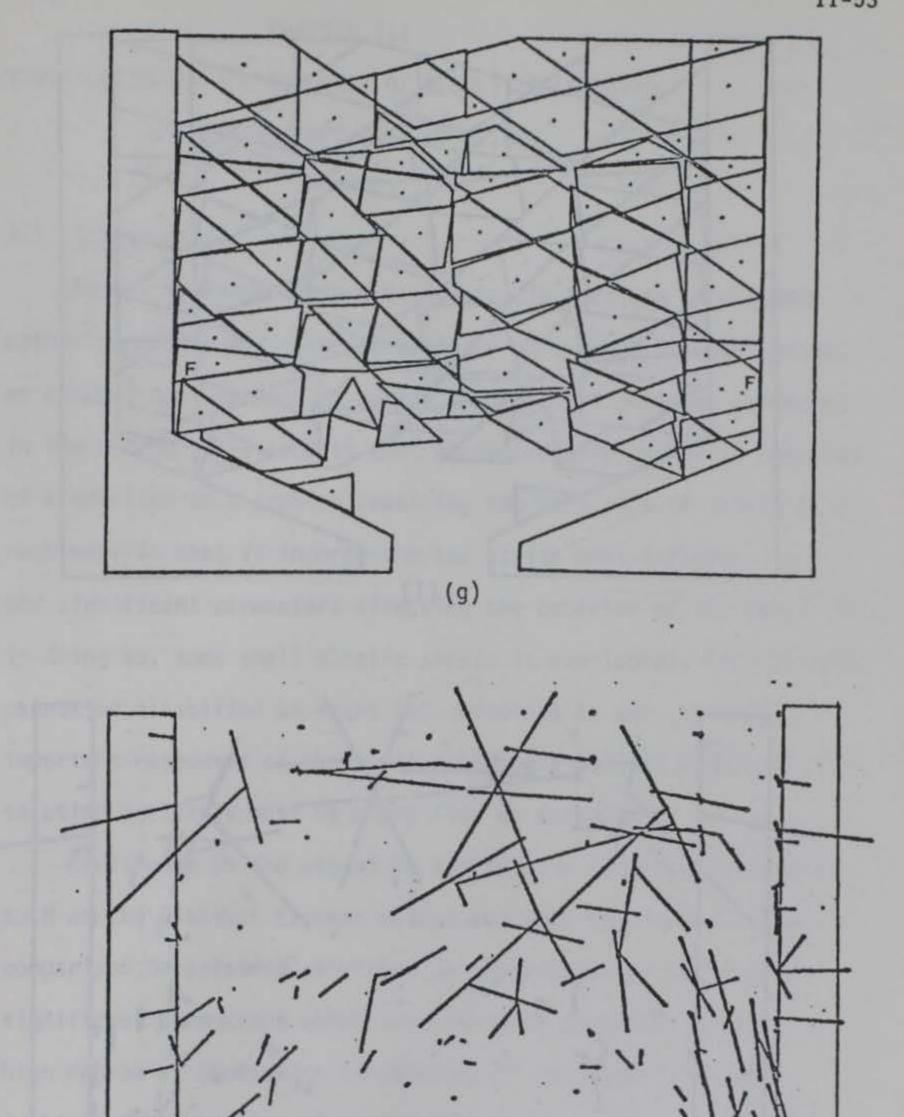


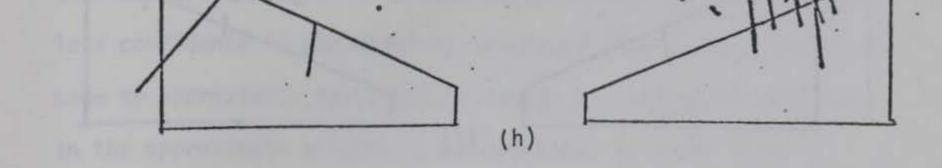


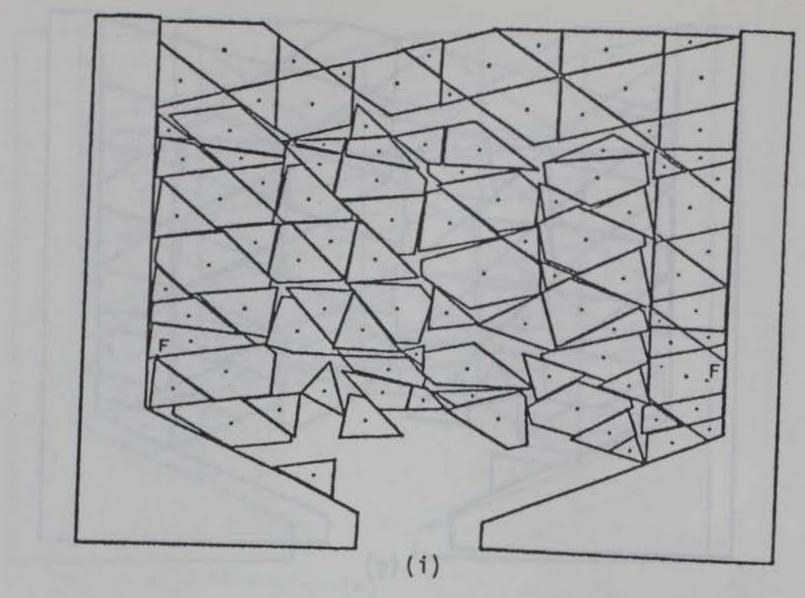


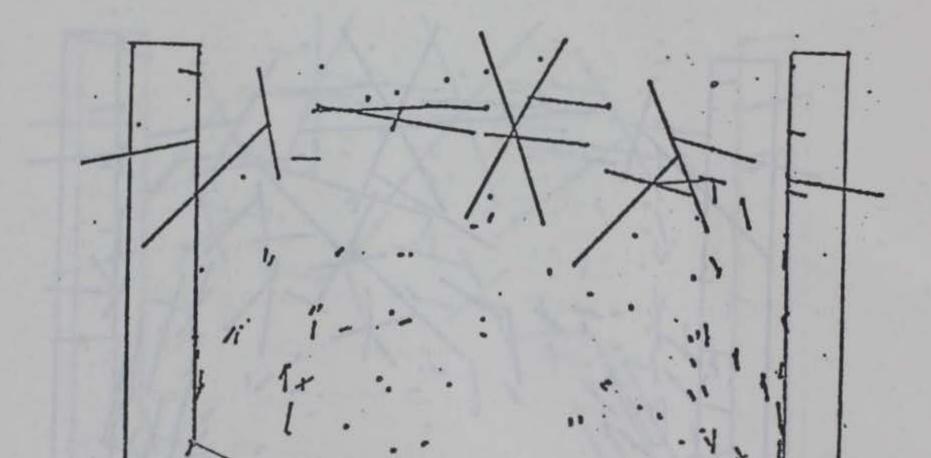


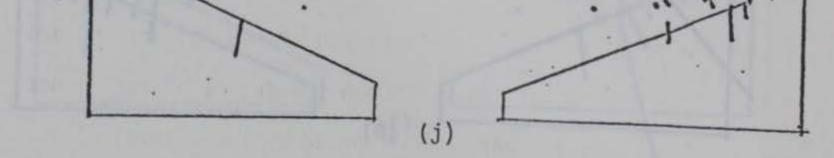












CHAPTER III VERIFICATION OF THE ACCURACY OF RESULTS CALCULATED

BY THE DISTINCT ELEMENT METHOD

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3.1 Introduction

As the Distinct Element method is, in fact, an approximate method to obtain the response behavior of a block jointed system, an attempt must be made to verify that the calculations performed in the method yield results that are acceptable. What is required of a solution to a problem involving the inclusion of joints in a rock mass is that it incorporate and assign most influence to the significant parameters affecting the behavior of the mass. If in doing so, some small elastic strain is overlooked, the solution cannot be classified as exact but, needless to say, if the important responses of the block system are modeled correctly, the solution certainly must be classified as acceptable.

Confidence in the use of an approximate numerical technique such as the Distinct Element method can best be developed through comparison to existing solutions to problems which include the significant parameters which the numerical technique models. A high degree of confidence is obtained if the numerical model duplicates the results of proven analytical solutions. Somewhat

less confidence in the model is developed if the comparisons are made to approximate solutions, although the degree of confidence in the approximate solutions, as evidenced by their level of acceptance by practicing engineers and designers, obviously must

be considered in the comparisons.

The problem of verifying the accuracy of solutions calculated by the Distinct Element method is compounded by the lack of analytical solutions that describe the behavior of a jointed rock mass. Instead, when dealing with the behavior of a jointed mass, most analytical solutions invoke approximations which draw upon empirically observed behavior models, soil mechanics theories and classical elastic solutions with the elastic parameters modified to reflect joint behavior. These types of models are severely limited in their applicability; for example, the elastic analyses are probably most valid for the case of very close jointing and the case of a very regular degree of jointing that can be characterized as an anisotropy. More general models for calculating the behavior of a jointed mass typically attack the problem by assuming simplified relationships between the parameters selected to typify the behavior. This type of model suffers in that the full implications of the roles these parameters play in the behavior of the mass are not yet fully understood.

What is needed then to perform a truly accurate comparison unfortunately does not exist. Rather, the very nature of the problem dictates that a choice be made between approximate techniques of analysis which often contain vastly simplified,

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empirically adjusted assumptions regarding the overall mass behavior which could possibly only be valid for a distinctly limited range of material properties. One group of approximate techniques, which is limited in its

scope to geometrically ideal problems, is acceptable for a comparison of this type. Limit Equilibrium solutions are concerned with the static equilibrium of bodies at the point of failure. Under this assumption, the frictional forces are assumed to be fully developed and thus force diagrams can be drawn and equilibrium equations written. This method requires the knowledge of the location of the failure surface and a minimal number of interacting blocks. Provided that the geometry of the mass can be represented simply, Limit Equilibirum principles are routinely used to calculate the response of a jointed mass.

In the sections that follow, five simple approximate models for the behavior of jointed masses are presented and the calculated responses are compared to that generated by the Distinct Element method. Included in these models are Limit Equilibrium analyses of: one block on an inclined plane with sliding and rotation possible; two interacting blocks, one in an active state, the other in a passive state; and, multiple interacting blocks both with and without the possibility of rotation. Also included are comparisons to physical models examined with a base friction apparatus, presented primarily for qualitative observations on the kinematics of large displacements, as well as a simple pressure distribution in a jointed mass where simplifying assumptions regarding material

behavior have reduced the problem to an application of the principles of static equilibrium. Common to the models chosen for comparison to the Distinct Element model are simple geometric properties and minimal assumptions regarding material behavior. As a result of this the models possess the additional feature that an intuitive insight into the ultimate response behavior is often possible. If it is possible to demonstrate that the simple models give the correct response, then it is much more meaningful if the Distinct Element model gives the same response.

3.2 The Base Friction Method

The base friction or base shear modeling technique is a physical, scale modeling technique described by Goodman (1972) that developed from the suggestion that the effect of gravity on a jointed rock slope could be simulated by shear forces on the base of the model as it was pushed over a plane surface. Alternatively, as in demonstrations attributed to Dr. E. Hoek (Goodman, 1976) the base may be moved while the model is restrained. The advantage of a horizontal assemblage of blocks lies in the fact that complex, unstable models may be constructed and failure observed as gravity is suddenly "switched on". Disadvantages arise due to the fact that accurate modeling of a real situation requires that a model material having the exact frictional properties of the real material must be found. In practice, exotic mixtures of flour, sand, salt and cooking oil are used to make a cuttable, semi-rigid modeling material. A material of this type has the advantage that discontinuities may be cut into it at arbitrary orientations; for the purposes of this investigation, however, as rigidity was of prime importance, 1 cm cubes of commercially available plexiglass were used to construct the models. The inability to orient discontinuities at arbitrary angles was not considered a severe liability in this investigation

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as the end result was simply to demonstrate qualitatively that the Distinct Element method would reproduce the expected modes of failure in several models where the failure modes were obvious. Figure 3.1 illustrates the small base friction apparatus used to study the behavior of the jointed models.

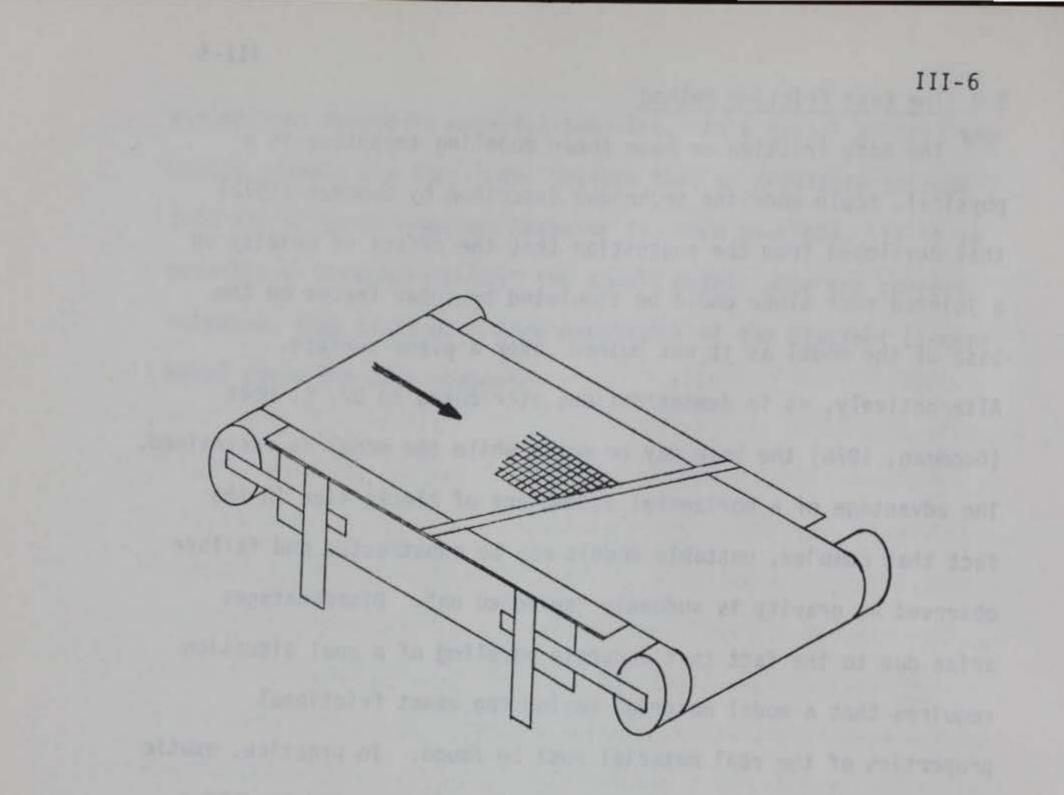


Figure 3.1 Diagramatic sketch of base friction apparatus used in comparison

Modeling techniques such as base shear are typically kinematic in that they reproduce the geometric features of the geologic structure and the excavation to a sufficient degree to establish possible modes of failure. However, they are not exactly scaled dynamically. For example, the base shear method does not give the correct response when a moving body acquires lateral momentum since in the base friction model, real accelerations are proportional to the driving belt velocity (Goodman 1976).

The implication of this is that in the absence of block to block contact, the only accelerations permitted in the model would be in the direction of the belt velocity as indicated in Figure 3.2. The Distinct Element model of this situation is included to demonstrate that momentum is indeed properly modeled.

However, several qualitative observations of a kinematic nature can be made: blocks which receive no supporting resistance must move downward under the effect of gravity; unconfined, geometrically unstable blocks must rotate and topple; and confined, geometrically unstable blocks must induce sliding in neighboring blocks as they rotate and topple. These three behavioral features of jointed systems can readily be simulated on a base shear apparatus by a laterally unsupported mine roof, an overhanging cliff and a cut slope in a jointed mass, respectively. These three failure models were chosen because, due to their simplicity, the kinematics of the failure are obvious. This makes them ideal for comparison with the Distinct Element method for it demonstrates that the Distinct Element method can calculate the proper failure mode for several situations for which the failure modes can be envisioned.

Figures 3.3, 3.4, and 3.5 illustrate a comparison of each of the three above mentioned failure modes by the base shear

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technique and the Distinct Element method. Little, if any, comment

appears necessary other than to point out the similarity of the

developing failure in all three cases.

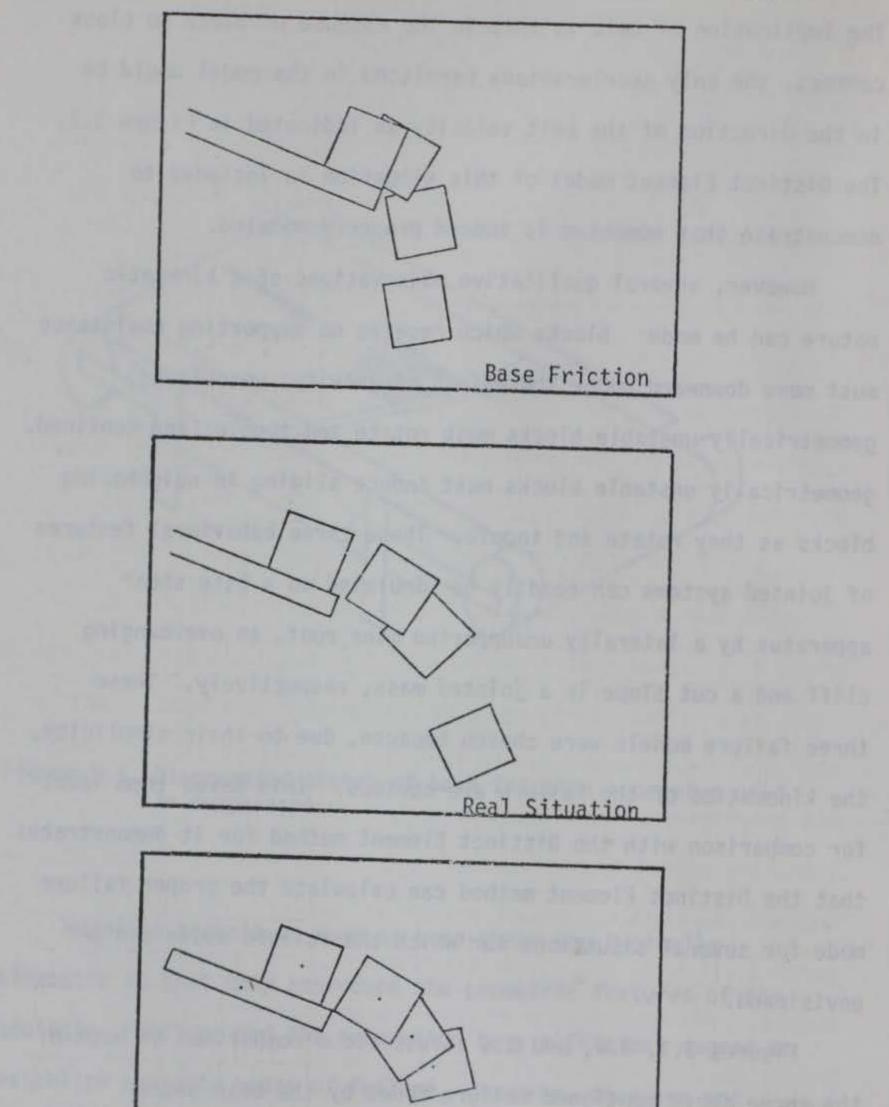
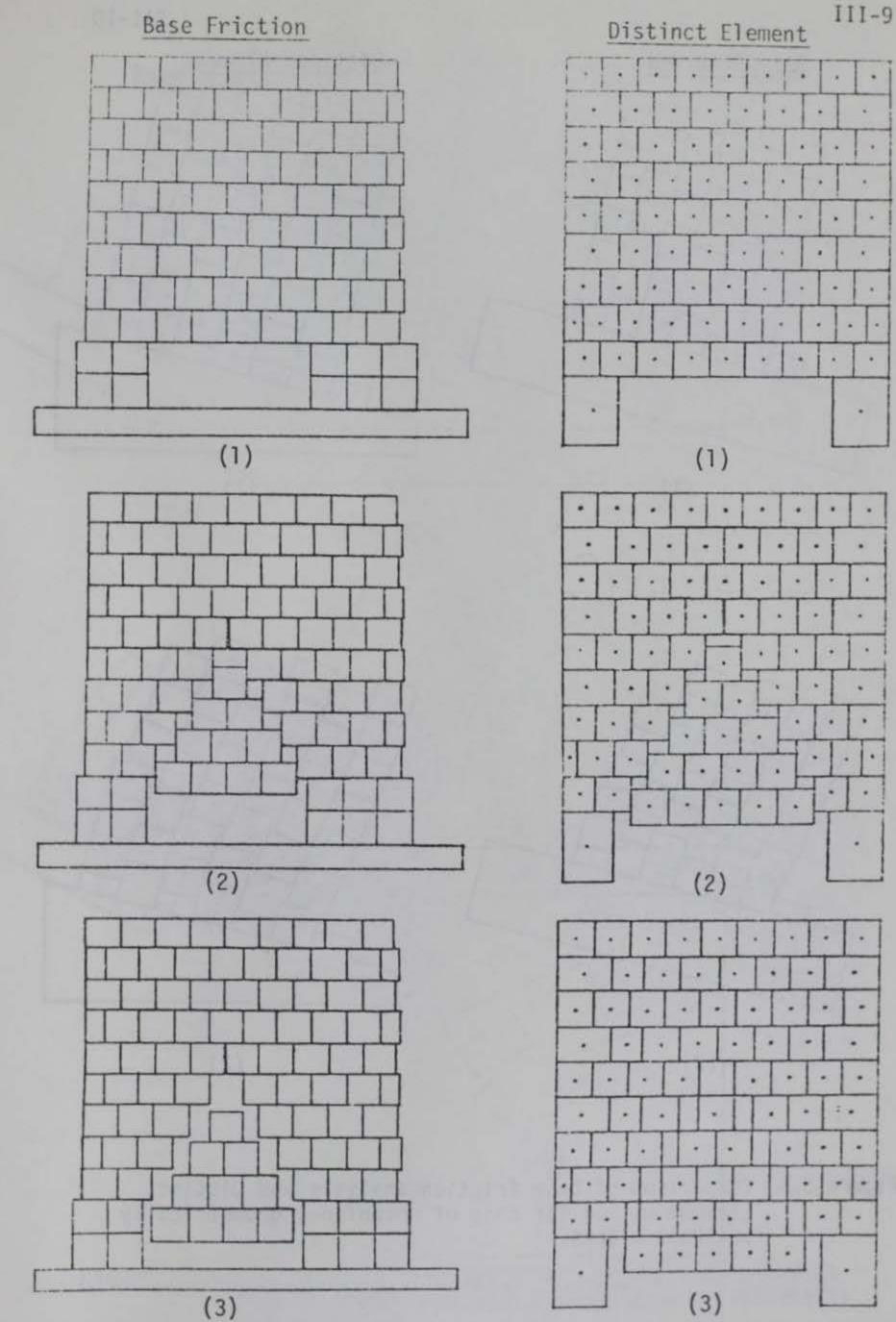


Figure 3.2 Dissimilarity of base friction model and Distinct Element method and real situation where momentum is not negligible.

Distinct Element



. $\left| \mathbf{g} \right|$.

Figure 3.3 Comparison of base friction analysis and Distinct Element method for case of unrestricted, gravity induced block displacement.



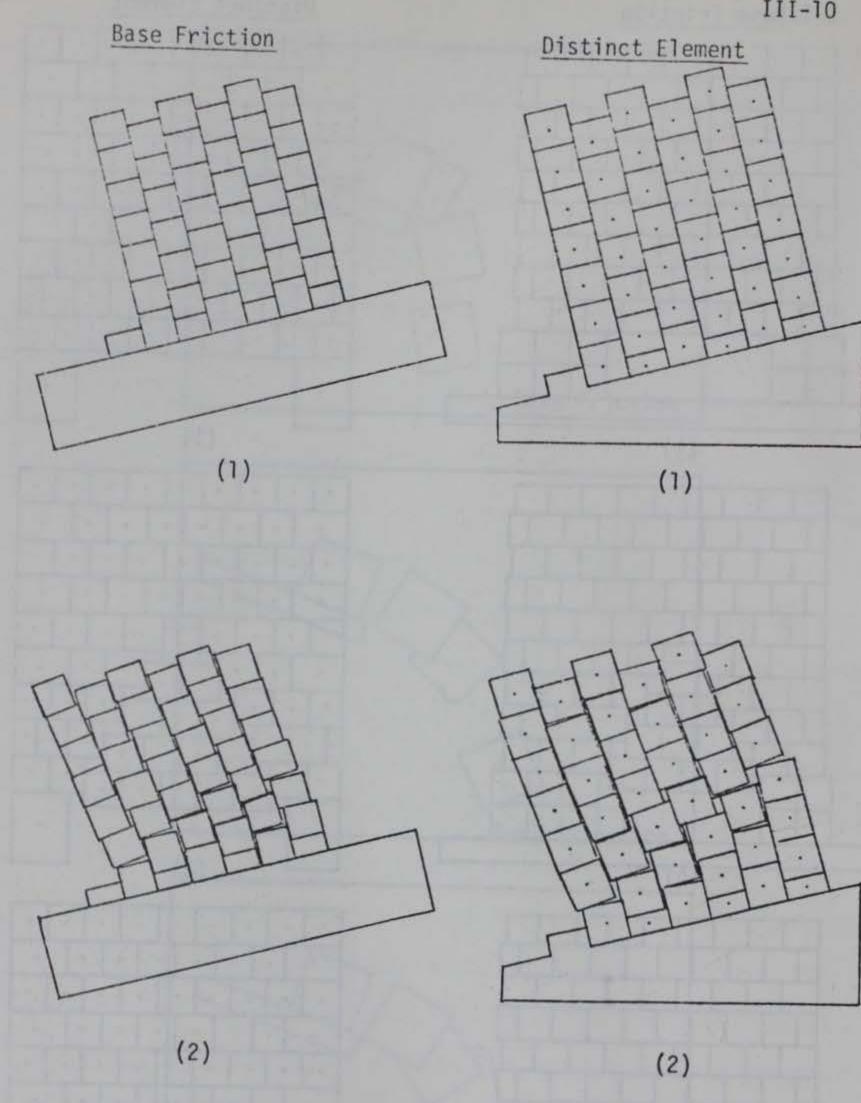


Figure 3.4 Comparison of base friction analysis and Distinct Element method for case of unconfined geometrically unstable blocks.

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Base Friction

Distinct Element

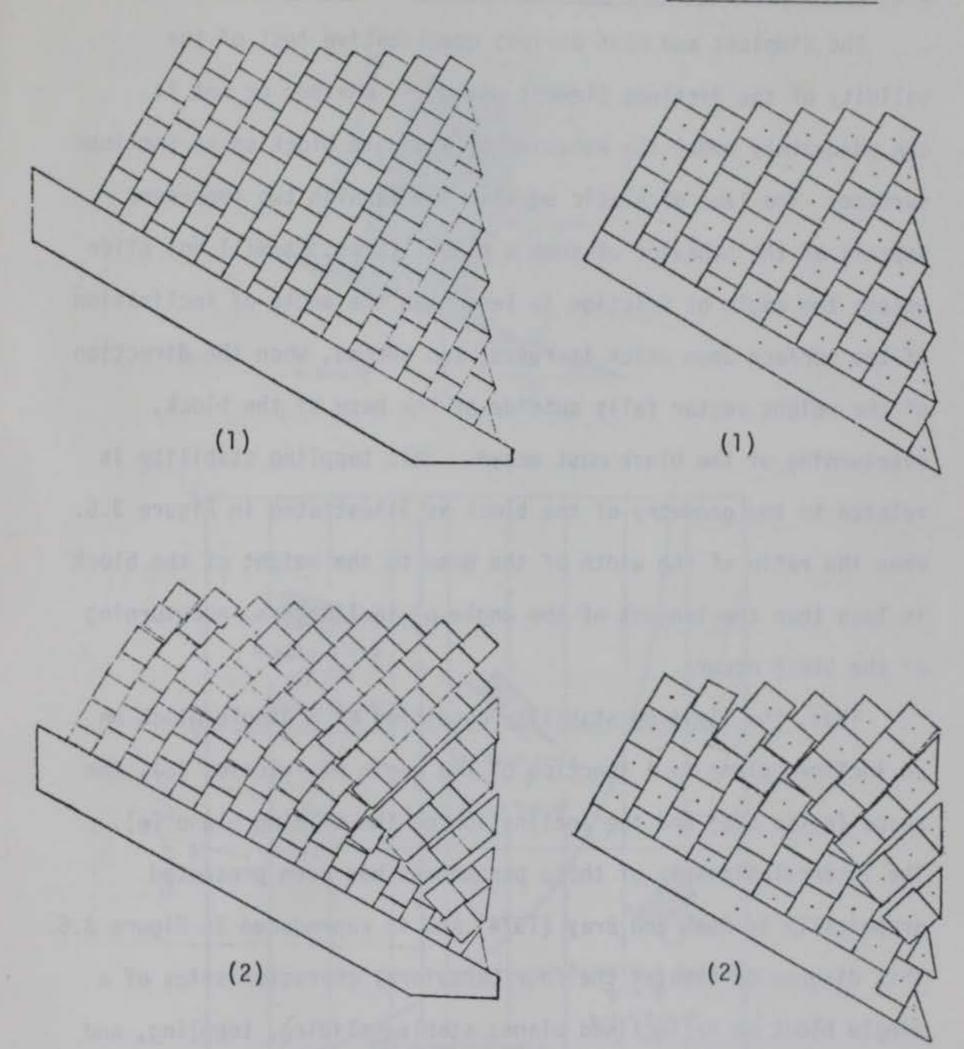


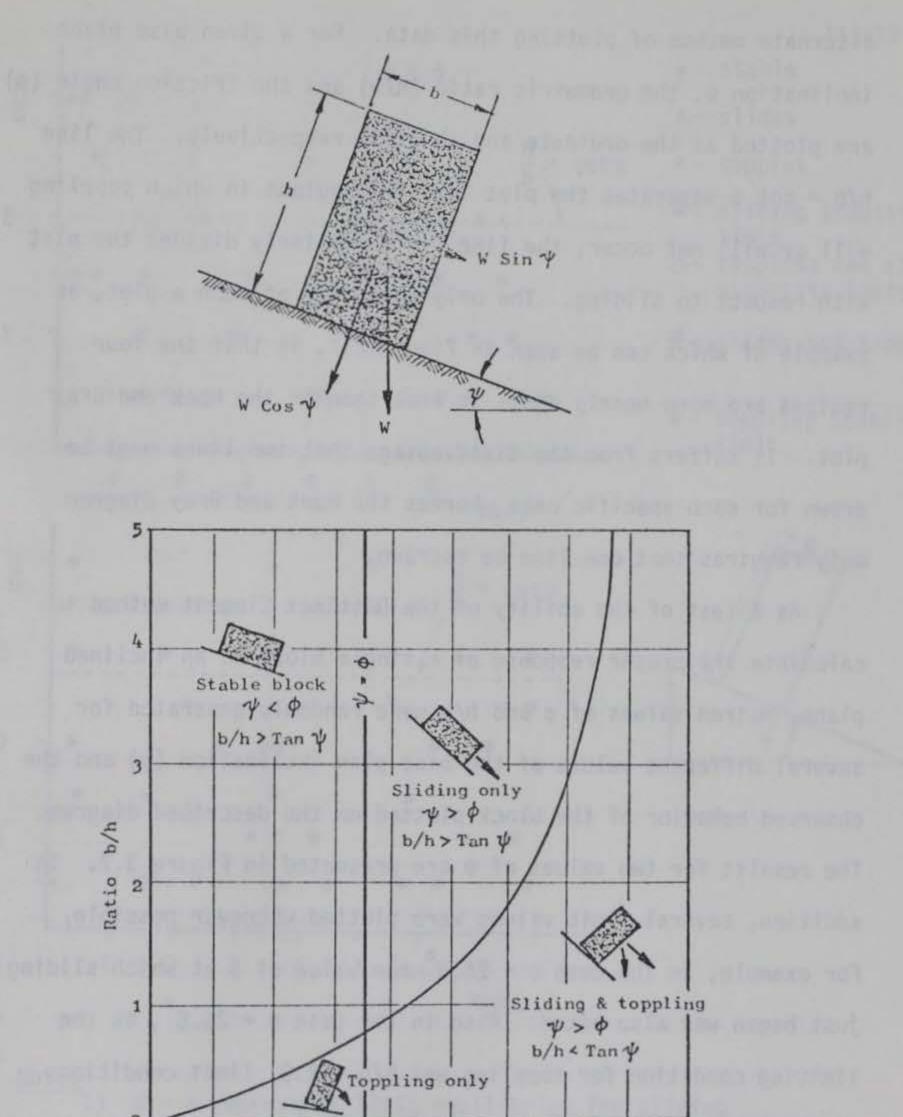
Figure 3.5 Comparison of base friction analysis and Distinct Element method for case of confined, geometrically unstable blocks.

3.3 Limit Equilibrium of a Single Block

The simplest and most obvious quantitative test of the validity of the Distinct Element method is whether or not it can adequately model the behavior of a single block on an inclined surface. The laws of static equilibrium furnish two important aspects of the behavior of such a block: first, it will not slide unless the angle of friction is less than the angle of inclination of the surface upon which it rests; and second, when the direction of the weight vector falls outside of the base of the block, overturning of the block must occur. This toppling stability is related to the geometry of the block as illustrated in Figure 3.6. When the ratio of the width of the base to the height of the block is less than the tangent of the angle of inclination, overturning of the block occurs.

Thus, the limiting stability condition of a single block on an inclined plane is a function of the angle of friction (ϕ), the shape (ratio h/b) and the inclination of the sliding plane (ψ). The interrelationship of these parameters has been presented graphically by Hoek and Bray (1974) and is reproduced in Figure 3.6. This diagram delineates the four behavioral characteristics of a single block on an inclined plane: stable, sliding, toppling, and a combination of sliding and toppling. Note that the line $\phi = \psi$ is not fixed on the diagram - it is moved laterally to specify the boundary for a given ϕ situation.

The line $\phi = \psi$ and the line h/b = cot ψ , representing limiting conditions for any specific block under consideration, suggest an



0 10 20 30 40 50 60 70 80 90

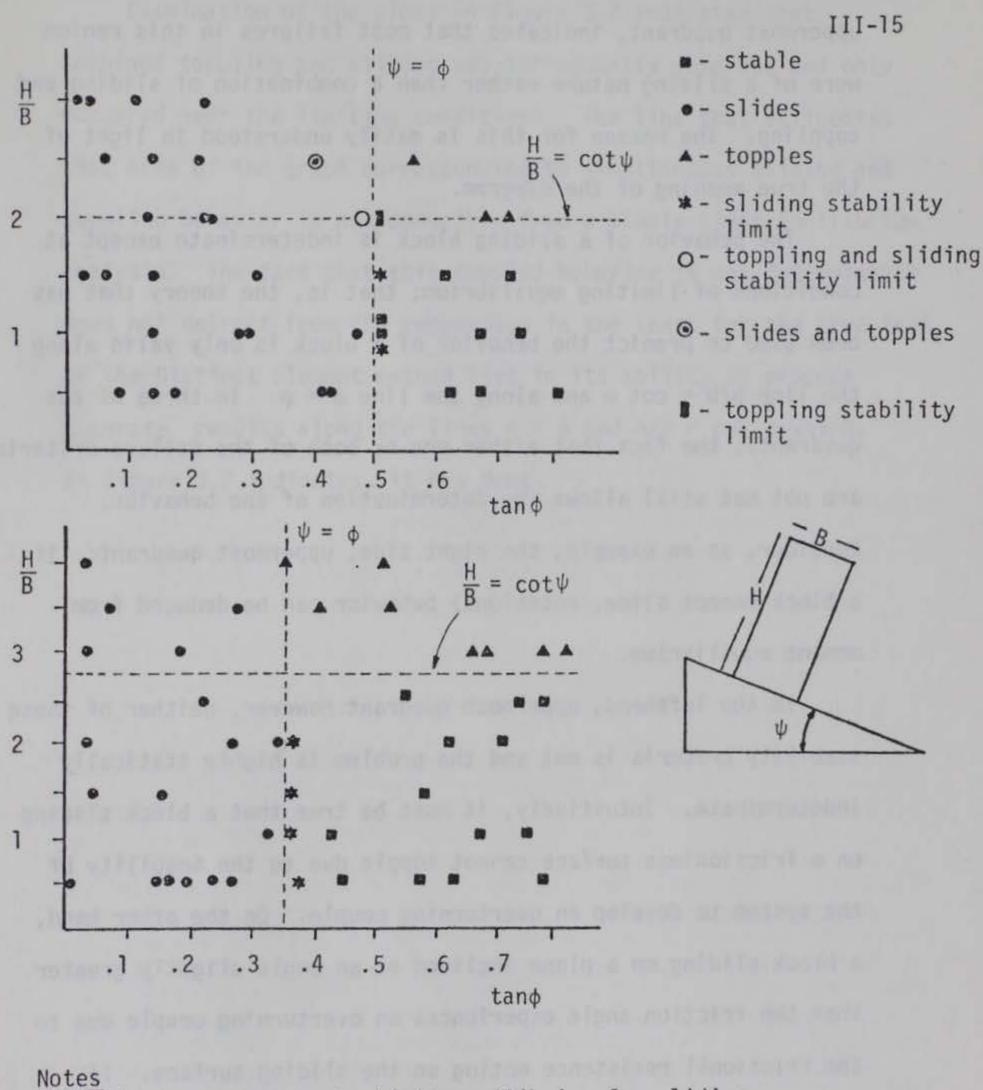
Base plane angle 🌾 - degrees

Figure 3.6 Conditions for sliding and toppling of a block on an inclined plane (from Hoek and Bray, 1974).

alternate method of plotting this data. For a given base plane inclination ψ , the geometric ratio (h/b) and the friction angle (ϕ) are plotted as the ordinate and abscissa respectively. The line h/b = cot ψ separates the plot into two regions in which toppling will or will not occur; the line $\psi = \phi$ similarly divides the plot with respect to sliding. The only advantage of such a plot, an example of which can be seen in Figure 3.7, is that the four regions are more nearly equal in area than on the Hoek and Bray plot. It suffers from the disadvantage that two lines must be drawn for each specific case whereas the Hoek and Bray diagram only requires that one line be redrawn.

As a test of the ability of the Distinct Element method to calculate the proper response of a single block on an inclined plane, paired values of ϕ and h/b were randomly generated for several different values of the base plan inclination (ψ) and the observed behavior of the block plotted on the described diagram. The results for two values of ψ are presented in Figure 3.7. In addition, several limit values were plotted whenever possible. For example, in the case $\psi = 26.6^{\circ}$ the value of ϕ at which sliding just began was also noted. Also in the case $\psi = 26.6^{\circ}$, as the limiting condition for toppling was h/b = 2.0, limit conditions

at which toppling just began were investigated. The results presented in Figure 3.7 show that the Distinct Element method is capable of accurately predicting the behavior of a single block on an inclined surface with respect to sliding or toppling failures. However, close examination of the left side,



1) $\psi = \phi$ represents limit equilibrium for sliding 2) H = B cot ψ represents limit equilibrium for toppling

Figure 3.7 Limit Equilibrium conditions for a single block on a plane surface: ϕ , H/B pairs randomly generated for constant ψ.

uppermost quadrant, indicates that most failures in this region were of a sliding nature rather than a combination of sliding and toppling. The reason for this is easily understood in light of the true meaning of the diagram.

The behavior of a sliding block is indeterminate except at conditions of limiting equilibrium; that is, the theory that has been used to predict the behavior of a block is only valid along the line $h/b = \cot \psi$ and along the line $\phi = \psi$. In three of the quadrants, the fact that either one or both of the failure criteria are not met still allows the determination of the behavior. Consider, as an example, the right side, uppermost quadrant: if a block cannot slide, rotational behavior can be deduced from moment equilibrium.

In the lefthand, uppermost quadrant however, neither of these stability criteria is met and the problem is highly statically indeterminate. Intuitively, it must be true that a block sliding on a frictionless surface cannot topple due to the inability of the system to develop an overturning couple. On the other hand, a block sliding on a plane inclined at an angle slightly greater than the friction angle experiences an overturning couple due to the frictional resistance acting on the sliding surface. If, additionally, the block geometry is conducive to toppling, then intuitively, the fact that the block is sliding should introduce an additional toppling moment. An analysis as simple as that illustrated in Figure 3.6 cannot predict the dynamic behavior just described as it is only concerned with limiting cases. Examination of the plots in Figure 3.7 indicates that combined toppling and sliding was infrequently observed and only occurred near the limiting conditions. The line that delineates that area of the graph corresponding to simultaneous sliding and toppling behavior is not deducible from a simple Limit Equilibrium analysis. The fact that this coupled behavior is not determinable does not detract from the comparison in the least for the true test of the Distinct Element method lies in its ability to produce accurate results along the lines $\psi = \phi$ and h/b = cot ψ which, as Figure 3.7 indicates, it has done.

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3.4 Two Block Limiting Equilibrium Model

Goodman (1976) presents a method by which a Limit Equilibrium analysis of two interacting blocks can be performed with the aid of a stereonet. Figure 3.8 illustrates the general nature of the problem; a rock slide consists of two free blocks, one of which is in an active or loading state, the other is in a passive or resisting state. Sliding of the passive wedge is initiated by load transfer from the active wedge which, by definition cannot be sustained by friction alone along its base planes; moment equilibrium is not considered.

The procedure consists of three steps:

- analyze active block with plane 3 as a free face: find F_p required
- analyze passive block with plane 3 as a free face, and with load - F_p
- 3. system is safe if resultant or passive block falls

within the friction cone to the normal to plane 2 Note that if the angle that the resultant on plane 2 makes with the normal to plane 2 is taken as the friction angle on plane 2, then limiting equilibrium conditions exist throughout the mass.

Several different geometries were analyzed by this method for comparison with the Distinct Element method. Care was taken to

ensure that the geometries chosen for analysis would fail with a minimal amount of rotation and with full frictional resistance developing on all planes in accordance with the basic theory. The results of several of the test cases are presented in Table 3.1, some of the geometries and the associated stereographic projections are presented if Figure 3.8.

The difference in the friction coefficient for stability on Plane 2 as calculated by two block Limit Equilibrium as compared to that calculated by the Distinct Element method was found typically to be on the order of one percent.

	Limit Equilibrium		Distinct Element		Relative Difference
Case	φ	μ	ф	μ	in µ
1	23.0 [°]	0.425	23.3°	0.430	1.2%
2	25.5°	0.477	25.7°	0.482	1.0%
3	30.6°	0.591	30.8°	0.597	1.0%
4	33.0°	0.649	33.1°	0.652	0.5%
5	37.6°	0.770	37.5°	0.767	-0.4%

Table 3.1 Comparison of the coefficient of friction required for stability as calculated by Limit Equilibrium and by the Distinct Element method.

Other geometries, in which rotation played a major part in the failure, were analyzed and compared by the two methods. A typical geometry investigated is illustrated in Figure 3.10. The friction coefficient calculated by two block Limit Equilibrium for this

geometry was found to be 0.554; the friction coefficient calculated by the Distinct Element method was found to be 0.490. The resulting difference in the friction coefficient was thus eleven percent. If, however, a Limit Equilibrium analysis

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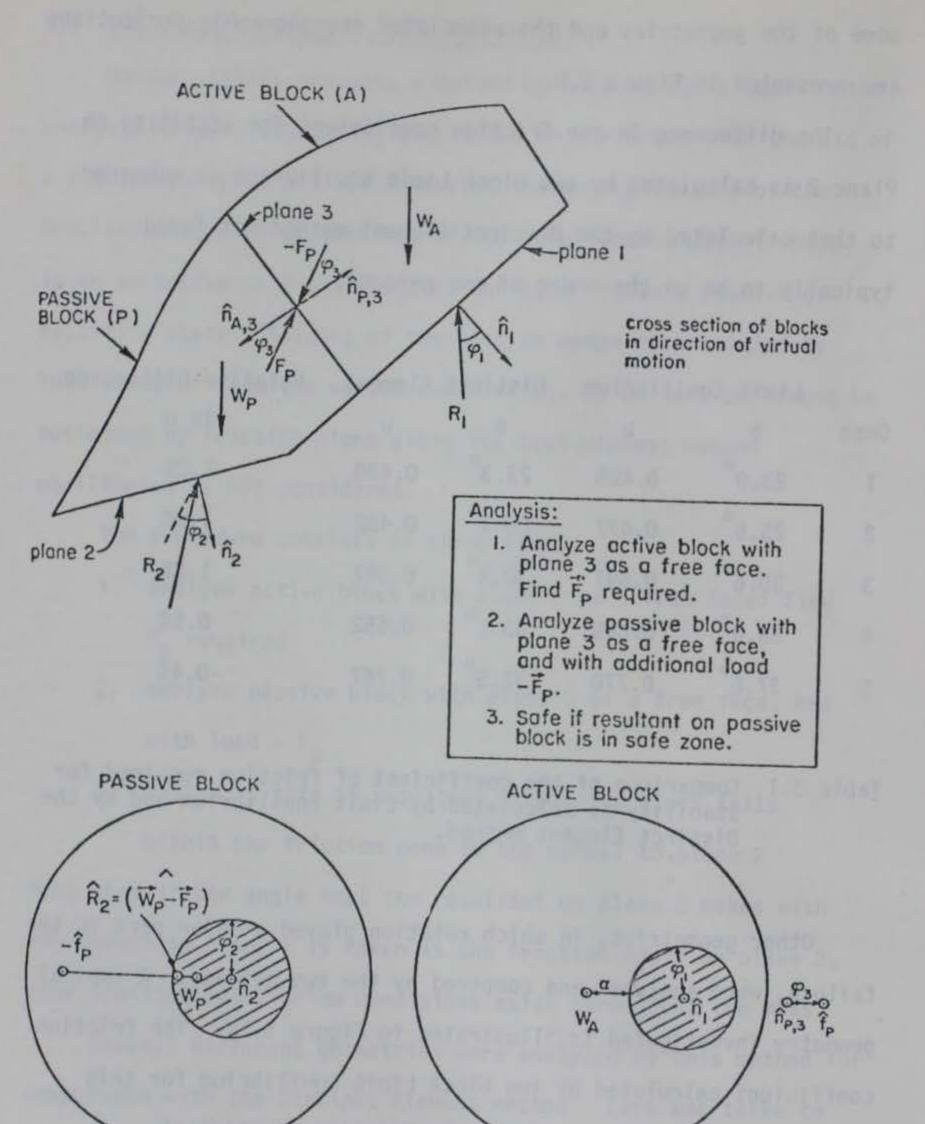
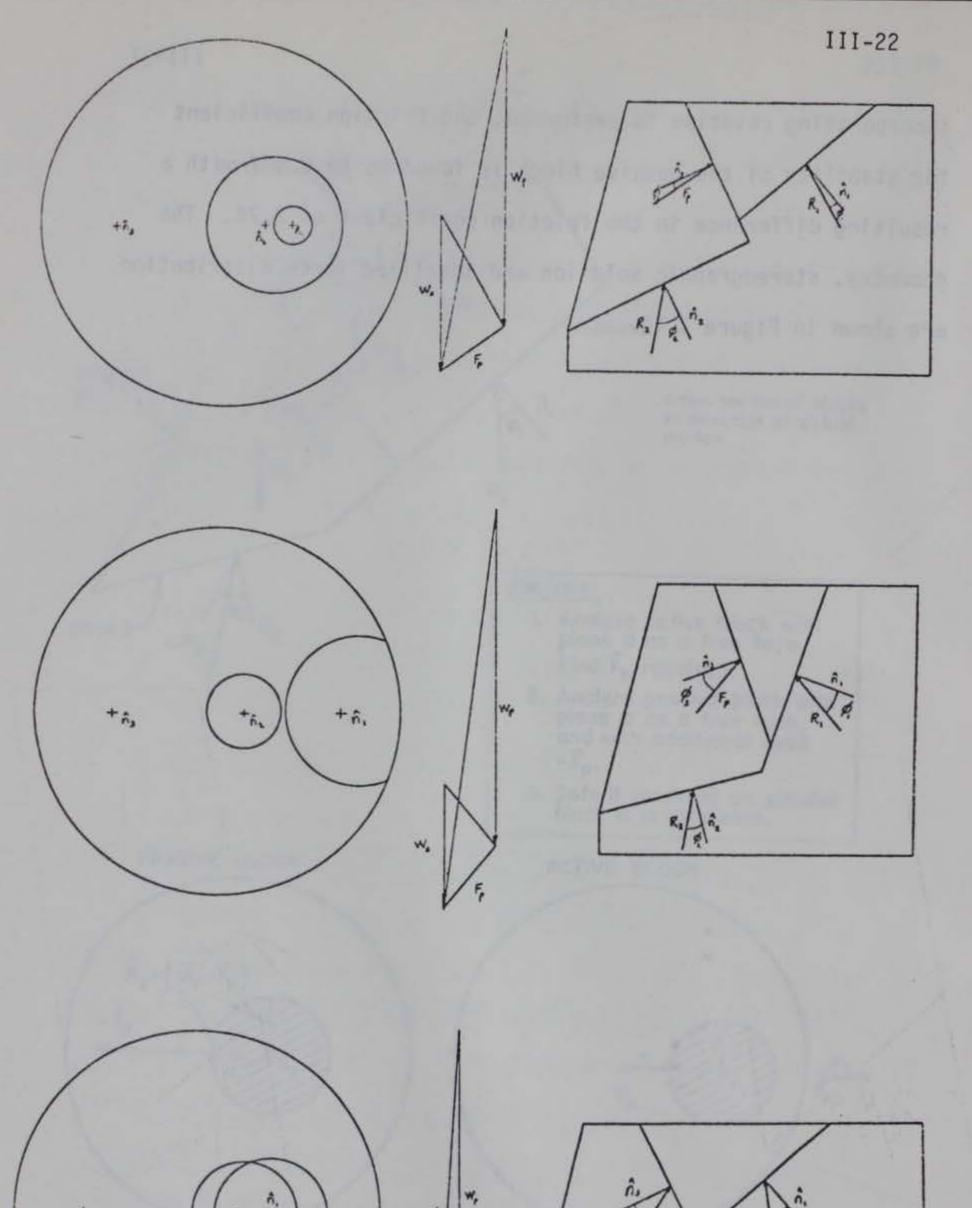


Figure 3.8 Parameters for two dimensional, two block Limit Equilibrium analysis (from Goodman, 1976)

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incorporating rotation is performed, the friction coefficient for stability of the passive block is found to be 0.477 with a resulting difference in the friction coefficient of 2.7%. The geometry, stereographic solution and idealized force distribution are shown in Figure 3.10.



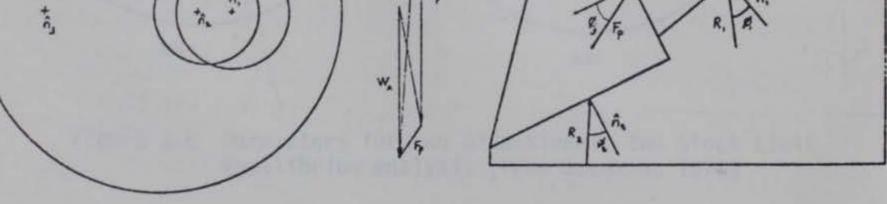


Figure 3.9 Geometries, force polygons and stereographic solutions for representative two block cases analyzed by Limit Equilibrium.

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WA \

(d)

(b) Figure 3.10 (a) (b) (c) Limit Equilibrium analysis of a two block model where toppling is an expected failure mode; (d)

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model where toppling is an expected failure mode; (d) Alternative force distribution for consideration of moment equilibrium.

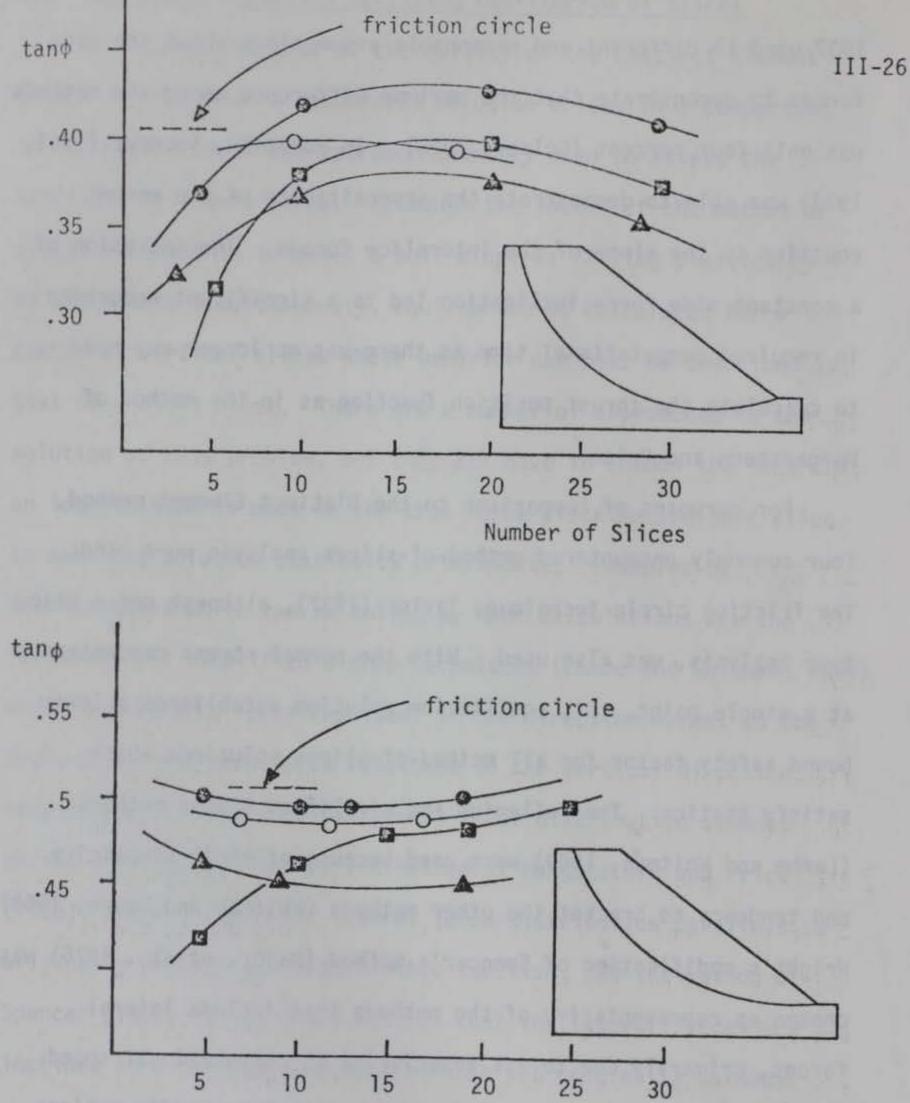
3.5 Embankment Stability Utilizing Equilibrium of Slices

An interesting test of the ability of the Distinct Element method to calculate a comparable solution arises in a comparison to the method of slices approach commonly used to assess the stability of a soil slope. Although the intent of the method of slices approach is to model a soil slope as failing plastically at all points simultaneously, equilibrium is calculated for a number of vertical slices whose behavior can best be described as that of a rigid block. There are a number of approaches to the solution of this problem, but they all have in common the fact that an idealization is made in the true force distribution on a slice to make the solution statically determinate. Examples of idealizations which can be solved by hand calculations are the Fellenius and simplified Bishop techniques (Lambe and Whitman, 1969) which assume zero force resultant in the direction normal to the failure arc and zero force resultant in the vertical direction, respectively. More complex lateral force distribution schemes exist, and are typified by the method of Morganstern and Price (1965), which assumes the lateral force distribution parallels an originally unknown but determinable function, and the method of Spencer (1967, 1973), which assumes that the lateral forces are inclined at a constant and determinable yet originally unknown

angle. The solution of these more complex schemes is typically highly iterative and best handled by a computer. To keep a proper perspective it must be noted that Fellenius chose to ignore the side forces in his method since the error introduced was on the order of five percent and that Beichmann in 1937 used 13 different and reasonable assumptions about the side forces to demonstrate that the maximum difference among the methods was only four percent (Golder, 1972). In addition, Spencer (1967, 1973) was able to demonstrate the insensitivity of the moment equation to the slope of the interslice forces. The inclusion of a constant side force inclination led to a significant reduction in required computational time as there was no longer any need to calculate the thrust position function as in the method of Morganstern and Price.

For purposes of comparison to the Distinct Element method, four commonly encountered method-of-slices analysis were used. The friction circle technique, Taylor (1937), although not a slice type analysis, was also used. With the normal stress concentrated at a single point, this equilibrium solution establishes a lower bound safety factor for all method-of-slices solutions which satisfy statics. The Fellenius and simplified Bishop methods (Lambe and Whitman, 1969) were used because of their simplicity and tendency to bracket the other methods (Whitman and Moore, 1963). Wright's modification of Spencer's method (Major, et al., 1976) was chosen as representative of the methods that include lateral forces, primarily due to its superiority in computational speed.

The results of the comparisons for two slope configurations are presented in Figure 3.11; the significant difference between the cases is that case B is more nearly planar owing to the larger radius of the failure surface. Inspection of the figure illustrates several interesting points as outlined in the following



Number of Slices

- Fellenius
 O Spencer Wright
- ▲ Simplified Bishop Distinct Element

Figure 3.11 Stability analysis by method-of-slices techniques and Distinct Element method.

paragraph.

Firstly, the variation in the friction coefficient required for Limit Equilibrium conditions is a function of the number of slices; the fact that Spencer's method, which utilizes lateral forces, is less sensitive to this parameter probably indicates the reason for this. As the blocks get thinner, they become rotationally unstable and lateral forces are required to maintain equilibrium. On the other hand as the number of slices becomes smaller, the system begins to act as an active/passive block system and once again, lateral forces are required for equilibrium to be reached. In practice, it is recognized that these problems are avoided if the number of slices is in the range of from ten to twenty. Within this range the friction coefficient as calculated by the Distinct Element method is within two percent of the method incorporating side forces (Spencer-Wright) and typically within five to seven percent of that given by either Fellenius or Bishop. Secondly, the friction coefficient calculated by the Distinct Element method diverges from that calculated by the other methods for a small number of slices. This is probably due to the fact that the Distinct Element method approximates the circular failure arc by a series of straight line segments and the possibility that any given segment could have an unwarranted influence on the

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sliding behavior. A given line segment could lower the inclination of the failure surface at any point along the slope with a corresponding decrease in the resultant friction coefficient required for stability. In contrast to this is the case where the failure arc is approximated by a larger number of slices; in this case the average slope of the failure arc is correctly represented. These two cases are illustrated in Figure 3.12.

Case A Case B

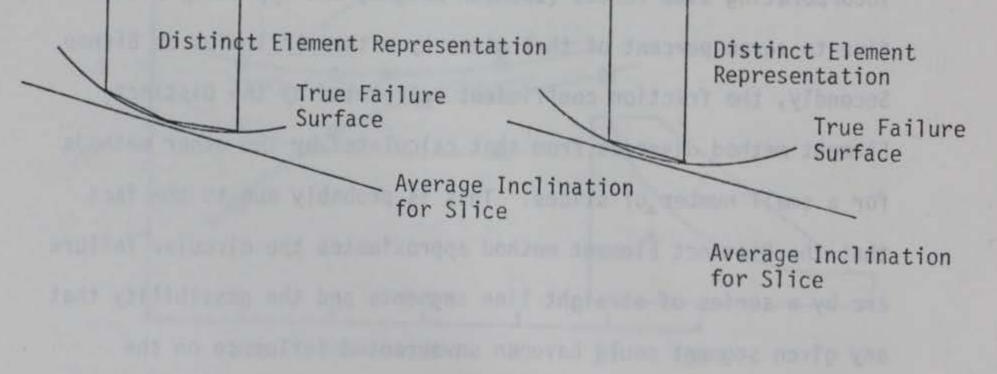


Figure 3.12 Possible mechanism (exagerated view) for divergence of Distinct Element method from slice methods as slice thickness increases. Note that in case A, sliding can occur on a line segment which has a higher inclination than the average for that section of the arc while this does not occur in case B.

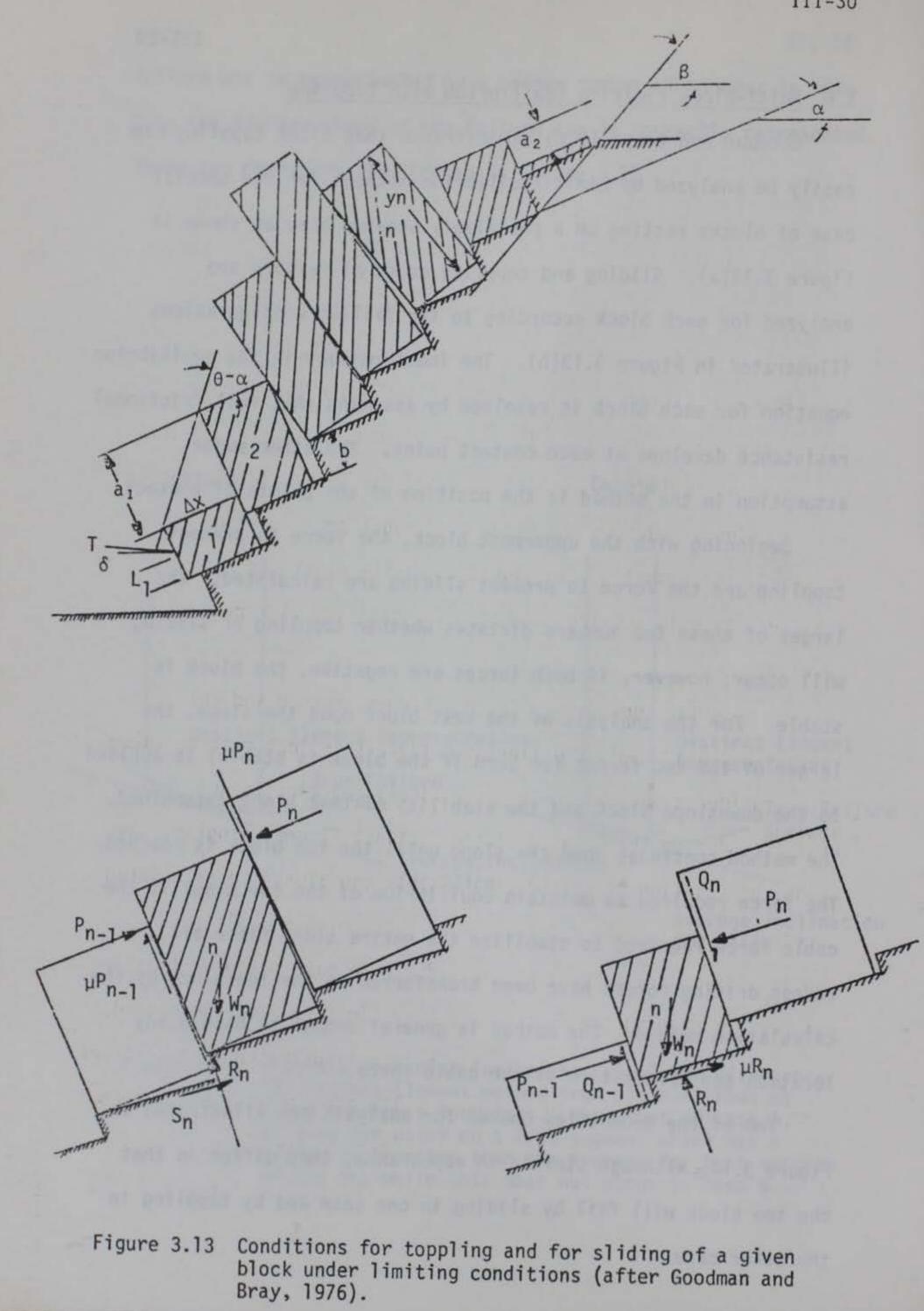
3.6 Multi-Block Limiting Equilibrium with Toppling

Goodman and Bray (1976) demonstrated that block toppling can easily be analyzed by Limit Equilibrium methods for the special case of blocks resting on a positively stepped base as shown in Figure 3.13(a). Sliding and toppling modes of failure are analyzed for each block according to the failing configurations illustrated in Figure 3.13(b). The indeterminacy in the equilibrium equation for each block is resolved by assuming that full frictional resistance develops at each contact point. The other major assumption in the method is the position of the points of contact.

Beginning with the uppermost block, the force to prevent toppling and the force to prevent sliding are calculated. The larger of these two numbers dictates whether toppling or sliding will occur; however, if both forces are negative, the block is stable. For the analysis of the next block down the slope, the larger of the two forces (or zero if the block is stable) is applied to the downslope block and the stability of that block determined. The method continues down the slope until the toe block is reached. The force required to maintain equilibrium of the toe block is the cable force required to stabilize the entire slope since all excess driving forces have been transferred to the toe block by the

calculation method. The method is general enough to handle any location and orientation of the cable force. Two of the geometries chosen for analysis are illustrated in Figure 3.14; although similar in appearance, they differ in that the toe block will fail by sliding in one case and by toppling in the other case.

III-30



One additional point must be considered when the mode of failure is dominated by toppling. Whereas the stability of a system of sliding blocks may be analyzed with the Distinct Element method by beginning with a condition that is stable with respect to frictional sliding and reducing the friction coefficient until failure occurs, the situation that exists when toppling modes of failure are present is more complex. On the one hand, frictional resistance on the sides of the block and at the corner about which rotation is occurring cannot be fully developed unless rotation induced lateral movement has been allowed to occur between blocks. But on the other hand, once some rotation has occurred, the geometric configuration of the blocks is such that a higher force is required to maintain stability with respect to toppling.

In a comparison of the Distinct Element method and the Goodman and Bray Limit Equilibrium method, this fact must be taken into consideration. Since the significant coordinates are always available during the running of the Distinct Element program, the amount of rotation of an individual block can always be calculated at any time during the running of the program. In addition, a sensitivity analysis relating cable force to base plane inclination was performed using the Goodman and Bray Limit Equilibrium method. The variation of the step inclination illustrated in the figure does not represent an actual change in the geometry of the model but reflects the actual displacement of the blocks due to rotational movements in the Distinct Element model. The value of the cable

III-31

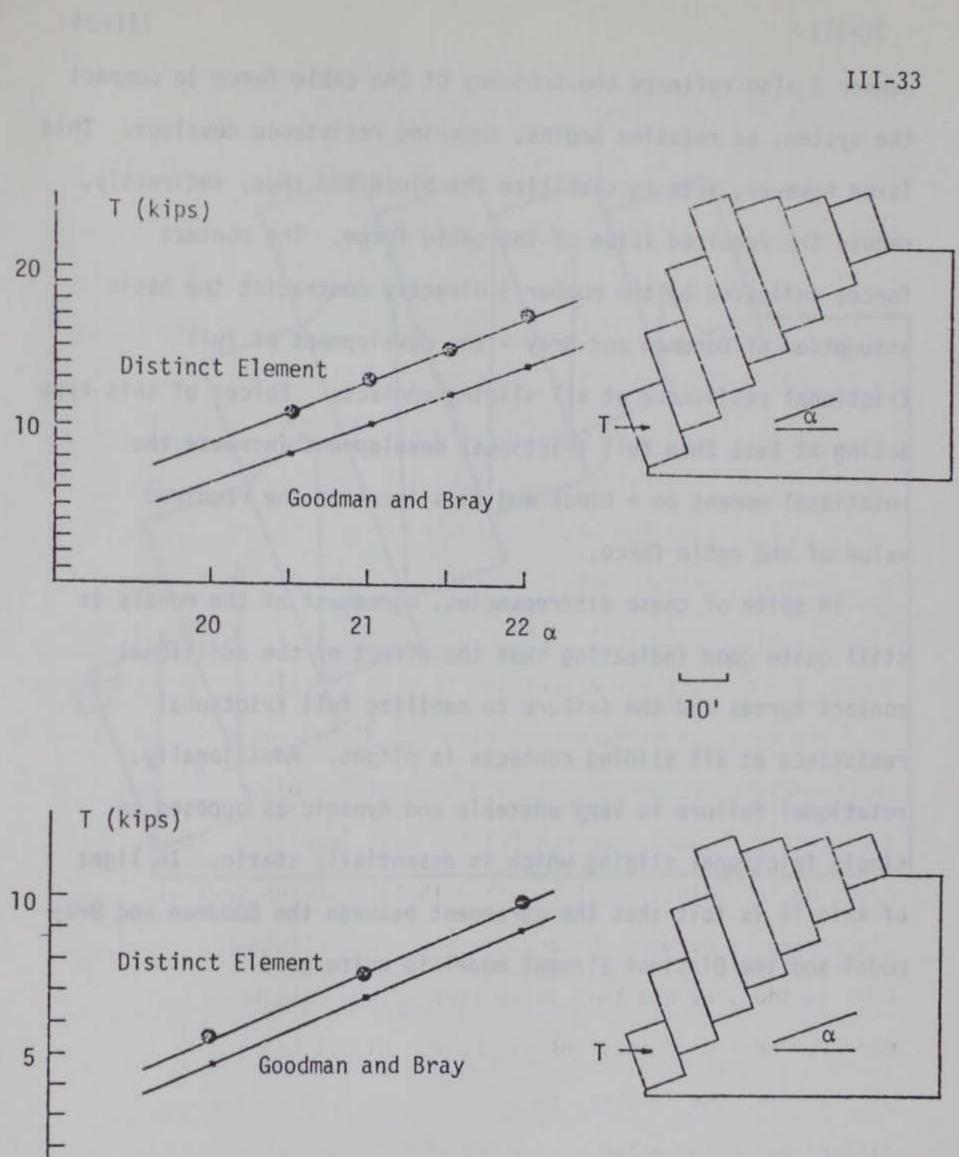
force determined by the Distinct Element method for several values of block rotation is illustrated. The corresponding values as determined by Goodman and Bray's method are also plotted for equivalent rotations. By comparing the data in this manner, there is assurance that the difference in calculated values is not due to a failure to compare equivalent models.

The results of the two comparisons are presented in Figure 3.14; part A illustrates the case of the toe block toppling and part B illustrates the case of the toe block sliding. Inspection of Figure 3.14 shows that the response of the Distinct Element model is similar to that of the Goodman and Bray Limit Equilibrium model; the cable force calculated is also similar for both models.

The relative difference in the calculated cable forces is approximately ten percent for the case of toe block sliding and approximately twenty percent for the case involving toe block rotation. Examination of Figure 3.15 illustrates several discrepancies between the contact force distribution assumed by Goodman and Bray and that calculated by the Distinct Element model. These discrepancies all have a direct bearing on the magnitude of the required cable force and help to explain the difference in the value of the cable force as calculated by the

two methods.

The contact forces indicated by the number 1 in the figure indicate "elastic" compression of the block system due to the applied bolt force and result in an increased value of the bolt force required for stability. The contact force indicated by the



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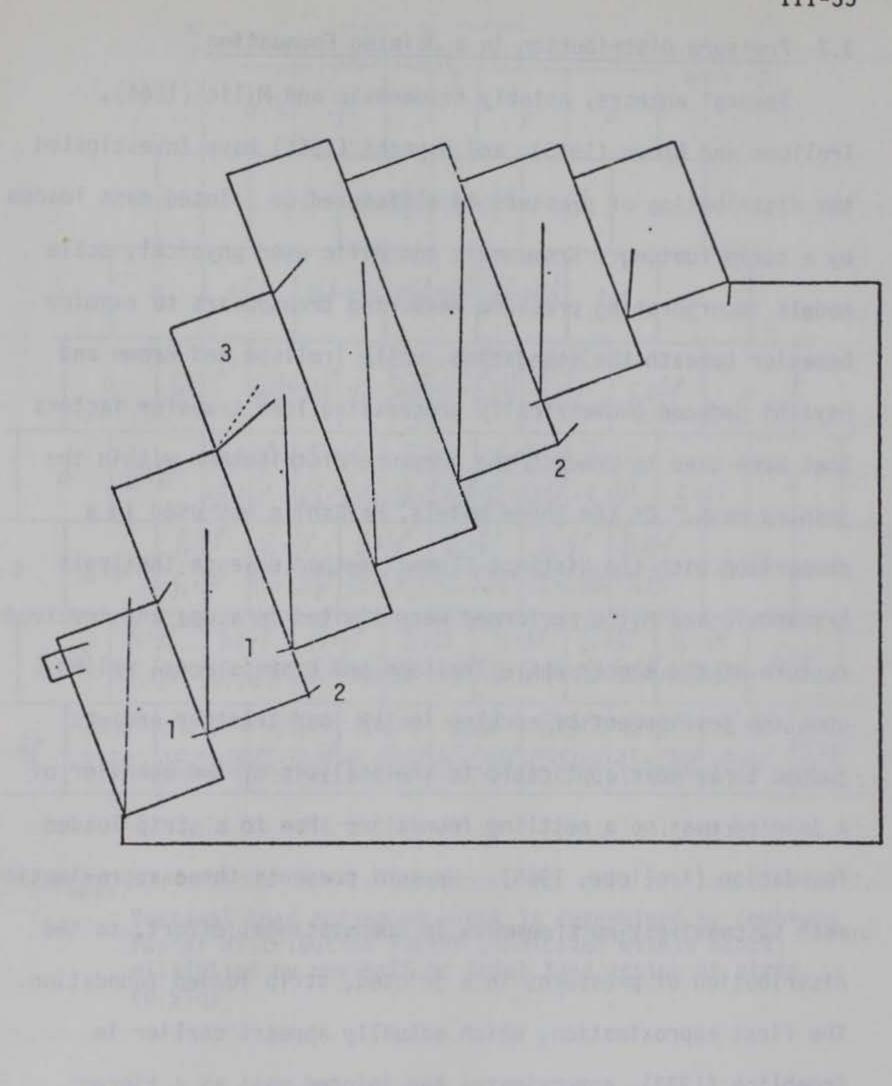
Figure 3.14 Comparison of Distinct Element calculated response of multi-block Limit Equilibrium and response as calculated by the method of Goodman and Bray (1976).

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number 2 also reflects the tendency of the cable force to compact the system; as rotation begins, shearing resistance develops. This force however, acts to stabilize the block and thus, indirectly, reduce the required value of the cable force. The contact forces indicated by the number 3 directly contradict the basic assumption of Goodman and Bray - the development of full frictional resistance at all sliding contacts. Forces of this type acting at less than full frictional development increase the rotational moment on a block and thus increase the required value of the cable force.

In spite of these discrepancies, agreement of the models is still quite good indicating that the effect of the additional contact forces and the failure to mobilize full frictional resistance at all sliding contacts is slight. Additionally, rotational failure is very unstable and dynamic as opposed to simple frictional sliding which is essentially static. In light of this it is felt that the agreement between the Goodman and Bray model and the Distinct Element model is quite good.





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Figure 3.15 Observed discrepancies in the contact force distribution assumed by Goodman and Bray (1976).

3.7 Pressure Distribution in a Jointed Foundation

Several authors, notably Krsmanovic and Milic (1964), Trollope and Brown (1965), and Hayashi (1966) have investigated the distribution of pressure in a fissured or jointed mass loaded by a strip footing. Krsmanovic and Milic used physical, scale models incorporating pressure measuring transducers to examine behavior beneath the foundation, while Trollope and Brown and Hayashi deduced geometrically progressing load transfer factors that were used to predict the pressure distribution within the jointed mass. Of the three models, Hayashi's was used in a comparison with the Distinct Element method because the tests Krsmanovic and Milic performed were limited in scope and involved rupture of the blocks while Trollope and Brown's model relied upon the development of arching in the load transfer and was judged to be more applicable to the analysis of the behavior of a jointed mass on a settling foundation than to a strip loaded foundation (Trollope, 1968). Hayashi presents three approximations, each successively more complex in computational effort, to the distribution of pressures in a jointed, strip loaded foundation. The first approximation, which actually appears earlier in Froehlich (1933), approximates the jointed mass as a tiered assemblage of point loaded simple beams; the resultant pressure

distribution for the case of no cohesion or frictional resistance reduces to the combined Pascal distribution as illustrated in Figure 3.16. The second approximation determines the elasticplastic boundary below which slip no longer occurs by means of the

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Note:

Vertical load acting on block is determined by combined Pascal distribution factor (indicated within block) miltiplied by one-half of total load acting on strip (0.5Tq)

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III-37

Figure 3.16 Hayashi's first approximation to the vertical, normal stress distribution in a fissured foundation combined Pascal distribution.

Boussinesq equations and the third approximation attempts to correct for the conversion of strain energy to heat as slipping occurs. As the second and third approximations introduce additional simplifying assumptions concerning the material behavior, the first approximation was chosen for the comparison with the Distinct Element method.

One of the resulting comparison plots is illustrated in Figure 3.17. Even plotted to an exagerated scale, the similarity is obvious. The maximum discrepancy in the two methods, relative to the total load, is seen to be only four percent. The dissimilarity in the two methods arises in Hayashi's failure to include rotational terms in his analysis. Examining the first row of blocks beneath the strip load shown in Figure 3.16 suggests that the central block, owing to a larger load, will undergo a slightly larger deflection than will the blocks on either side. This will result in an inward rotation of the two side blocks and a corresponding increase of load in the region beneath the central blocks. Following this line of reasoning it is easy to see that had Hayashi considered rotations in his model, the resulting pressure distribution would have been, from a qualitative viewpoint, slightly higher in the central region and lower on the sides bringing it more in line with the pressure distribution calculated

by the Distinct Element method.

III-39

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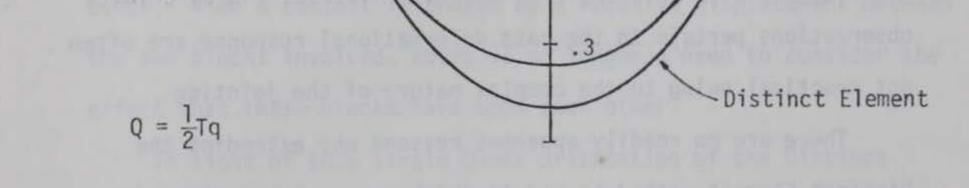


Figure 3.17 Vertical stress on a horizontal plane in a fissured foundation by the Distinct Element method and Hayashi's (1966) method.

3.8 Summary

It seems appropriate to conclude with a brief summary of the comparisons just presented, for the credibility of the remainder of this dissertation depends in part upon the acceptance of the validity of the Distinct Element method on the basis of the simple comparisons presented. Using a base shear apparatus, it was demonstrated qualitatively that the Distinct Element method calculated kinematically correct responses for several classes of complex problems where intuitive projections of the resultant mass deformational response were possible. For those Limit Equilibrium analyses of block models which represented essentially static situations, agreement was typically within one or two percent; even for the more dynamic situation involving multiblock rotations, agreement was on the order of ten percent. Finally, for that situation where it was possible to duplicate all of the assumptions regarding mass behavior, the Distinct Element method was observed to calculate a pressure distribution beneath a strip loaded foundation that was essentially similar to that calculated by Hayashi's (1966) theory.

Confidence in the method depends upon extending this credibility in the Distinct Element obtained solutions to problems where analytical solutions are not possible and where intuitive

observations pertain to the mass deformational response are often not practical owing to the complex nature of the jointing. There are no readily apparent reasons why extending the Distinct Element method to models which are more complicated geometrically should result in answers that are any less acceptable than those generated for the preceeding comparisons. The Distinct Element formulation contains no underlying requirements to dictate where failure surfaces should develop nor does it require that the failure mode must somehow be reducible to idealized mechanisms of arching, toppling, or sliding. No mass elastic response equations with empirically modified parameters are incorporated in the model; no "joint elements" need be formulated. In fact, owing to the explicit nature of the formulation there is not even a need to form a stiffness matrix relating block deformations to interblock loads.

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The Distinct Element formulation is oriented toward the behavior of each block as an individual mass. The kinematic behavior of each block is independently calculated using Newton's law of motion; each block senses the blocks surrounding it only as boundary conditions. If the movement of a block leads to penetration or relative movement along the surface of another block then the normal and shear stiffness will lead to interblock contact forces by a simple application of Hooke's law with an upper limit to the forces set by the Mohr-Coulomb relation. These forces are simply treated as boundary conditions for the first block. When a contact is broken by a relative displacement between the two blocks involved, there is no longer a need to consider the effect that these blocks have upon each other. In light of this single block orientation of the Distinct Element formulation there is no readily apparent reason why the only difference between a problem involving only a few blocks and

one involving tens or hundreds of blocks should be anything more than the extended time required to perform the calculations.

It should be noted, however, that the time step used in the calculation cycle is sensitive to the number of contact points a single block experiences at a given time. An increasing number of contact points can lead to numerical instabilities; this simply necessitates a reduction in the time step and is not an indication that the Distinct Element formulation is incapable of solving problems where single blocks simultaneously experience multiple contact points. In the present configuration, the equations are stable up to a maximum of eight points per block.

Additional verification comparisons of Distinct Element calculated responses are presented in the remaining chapters whenever it is possible to express quantitatively the behavior of the block jointed mass under consideration. The high degree of correlation exhibited by the comparisons presented in this chapter is also found to be true for the comparisons presented in the later chapters.

CHAPTER IV

THE STABILITY OF UNDERGROUND EXCAVATIONS IN JOINTED ROCK

4.1 Introduction

The first step in a rational support design method must logically be to predict whether or not a need for support actually exists. Rather than categorically stating that an excavation will or will not be stable if unsupported, it is more realistic to analyze a given situation by varying the values of the input parameters to determine those parameters to which the given excavation will be most sensitive. Using realistic values of the design parameters it can be determined if the excavation can be expected to stand unsupported or if support will be required. This type of investigation is typically found to be very sensitive to the input parameters, particularly those such as joint orientation and spacing, and the magnitude of the pre-existing stress field. Within the context of the expected variation of the parameters in the real situation it is then possible to make a qualitative statement about the stability of the excavation. This typically could be expressed in one of three ways: (1) within the expected variation of the input parameters the proposed excavation should be stable; (2) the expected variation in the input parameters indicates that

the excavation may or may not be stable, suggesting a possible need for light supports; or (3), realistic variation of the input parameters indicates that the excavation will not stand unsupported, suggesting the need for heavier supports. This chapter presents the results of numerous analyses of the

behavior of excavations in jointed rock in an attempt to determine which parameters had the greatest effect on the stability of the excavation. The models chosen for analyses are characterized by simple joint configurations and the behavior examined through the contact forces that exist between the blocks. This behavior is then interpreted in light of arching theory.

The term arch usually conveys the concept of a vaulted opening so that arching seems to describe the process by which the vaulted opening is formed. As used by Woodruff (1966), the term arching refers to the natural process by which a fractured material acquires a certain ability to support itself through the resolution of the vertical component of its weight into diagonal thrust. Arching theories examine the processes by which this stress transfer is accomplished.

Arching theories are based upon an analysis of beam behavior such as that presented by Woodruff (1966) which is illustrated in Figure 4.1(a). The analysis indicates that zones of tension and compression exist in the strata above the opening. In recognition of the fact that rock is relatively weak in tension, the lower row of the strata above the excavation is represented as being comprised of two independent blocks. The compressive forces which act to maintain the stability of the two blocks above the excavation are illustrated in Figure 4.1(b). The similarity of this force distribution to that of a three hinged structural arch is obvious; an analysis of excavation roofs in this manner is often termed linear arch analysis. As noted in Figure 4.1(b) no vertical force transmittal to the two roof blocks is assumed to occur. Thus linear arch analysis, in this simple form at least, is an analysis of the lower row of strata only.

A significant portion of the results of this chapter are based upon the recognition of arching patterns in the Distinct Element calculated contact force distributions in the jointed rock surrounding an excavation. It is worthwhile then to briefly describe the origin of the contact forces and the manner in which the arches are recognized.

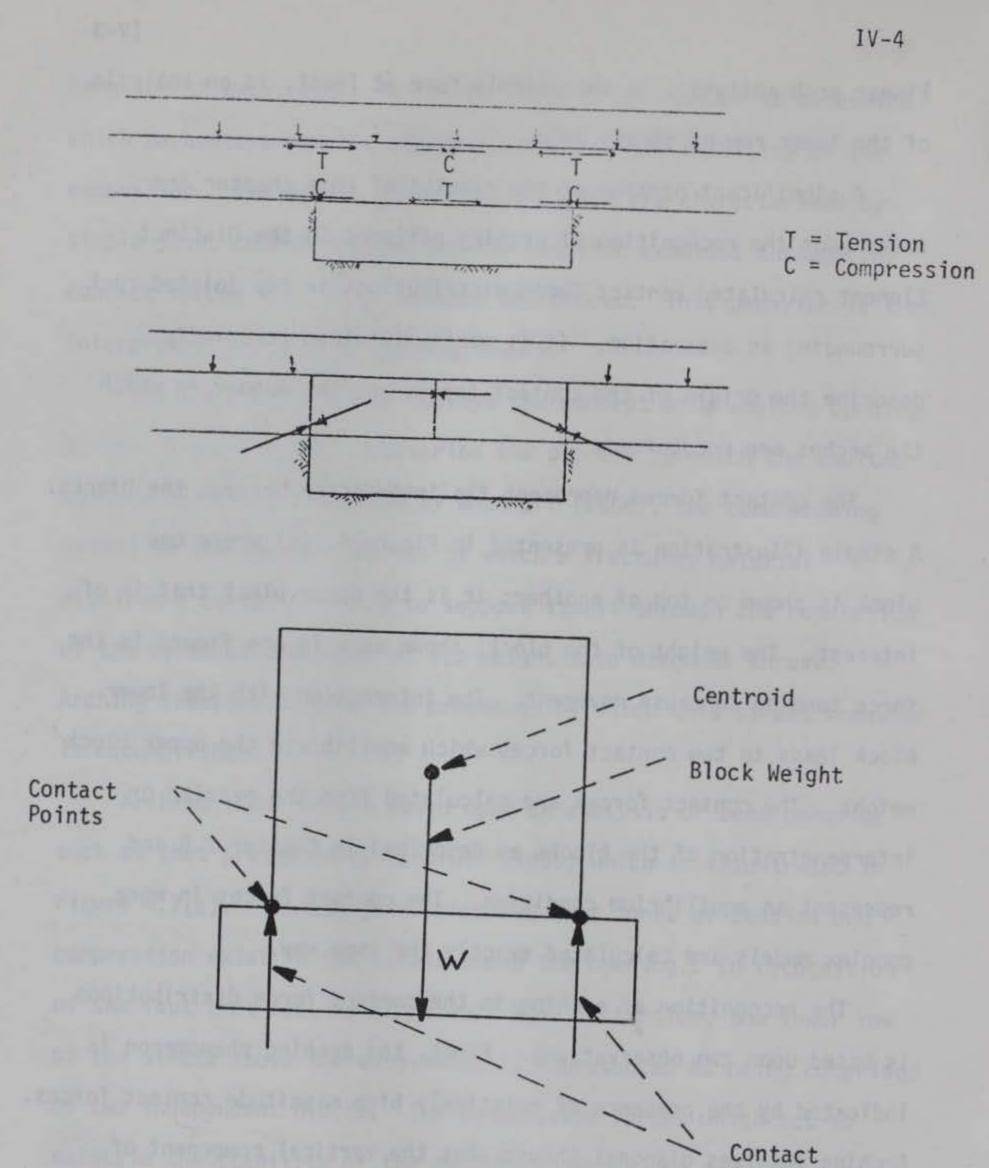
The contact forces represent the interaction between the blocks. A simple illustration is presented in Figure 4.1(c) where one block is shown on top of another; it is the upper block that is of interest. The weight of the block, shown as w in the figure is the force tending to cause movement. The interaction with the lower block leads to two contact forces which equilibrate the upper block weight. The contact forces are calculated from the overlap or interpenetration of the blocks as described in Chapter 2.8 and represent an equilibrium condition. The contact forces in more complex models are calculated exactly the same way.

The recognition of arching in the contact force distributions is based upon two observations. First, the arching phenomenon is indicated by the presence of relatively high magnitude contact forces. Arching involves diagonal thrust, but the vertical component of this thrust must be at least equal to the weight of the blocks being

supported by the arch action. Since the arch thrusts typically form

at low angles, the horizontal component of the thrust is usually

large. The recognition of arching also is based upon the necessary



Forces

Figure 4.1 (a) General distribution of stress in a beam over an opening; (b) self supporting linear arch model; and (c) contact forces due to weight of block.

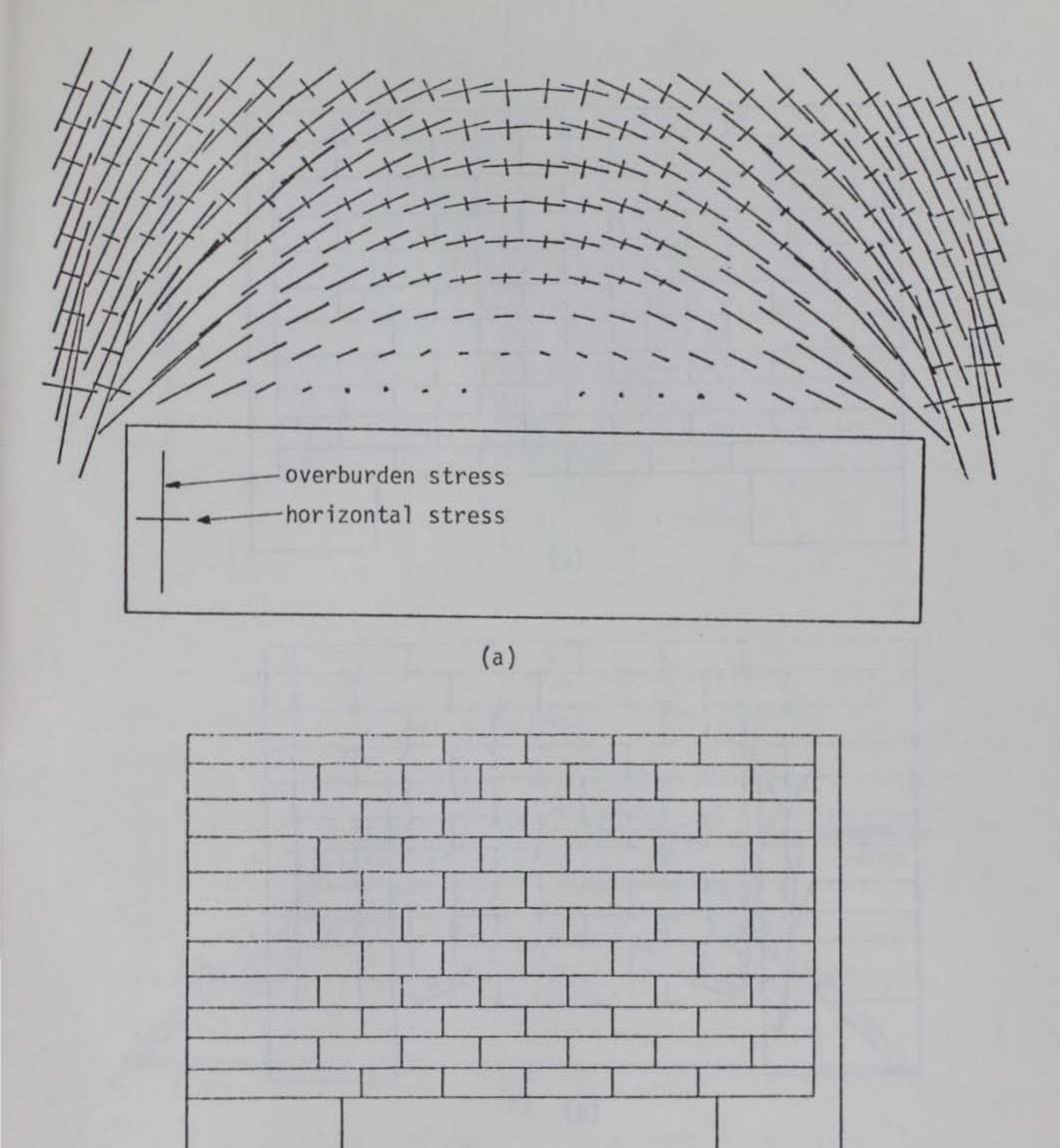
continuity of the force distributions. In particular, a block which is in equilibrium can have no unbalanced forces acting on it. Thus, the occurance of high contact forces in a region of low contact forces can only be possible if some mechanism is acting to transfer these forces to a high stressed region.

The analyses presented in this chapter indicate interactions exist within the mass which are typically neglected by arching theory. The analyses also indicate trends suggesting which input parameters have the most effect on the stability of an excavation in jointed rock.

4.2 General Observations on Force Distribution Around Excavations in Jointed Rock

An elastic analysis of the behavior of the rock surrounding an excavation invariably leads to the conclusion that the vertical stress component is transferred to the rock on either side of the excavation resulting in a region of relatively low stress immediately above the excavation. This fact has been demonstrated many times in the past by using photo elastic models and recently by using Finite Element analysis. A typical plot of stresses surrounding an opening in an elastic medium is presented in Figure 4.2(a). Note that a zone of tension exists at the crown.

The Distinct Element method can be used to study the redistribution of stress due to an excavation in a jointed medium. As an example, consider the model of the roof of an excavation presented in Figure 4.2(b). Owing to the discontinuous nature of the vertical jointing, only blocks in the lower four rows are able, from a kinematic standpoint, to move into the excavation. The weights of all of the blocks, drawn to a common scale, are illustrated in Figure 4.2(c). All of the contact vector distributions for the jointed models illustrated in Figure 4.2 utilize the same force scale. Figure 4.2(d) illustrates the redistribution of forces that occurs as the room is excavated. Analogous to the elastic model, the bulk of the stress is transferred to the material on either side of the excavation and a destressed, triangular zone is seen directly above the opening. The lower portion of the

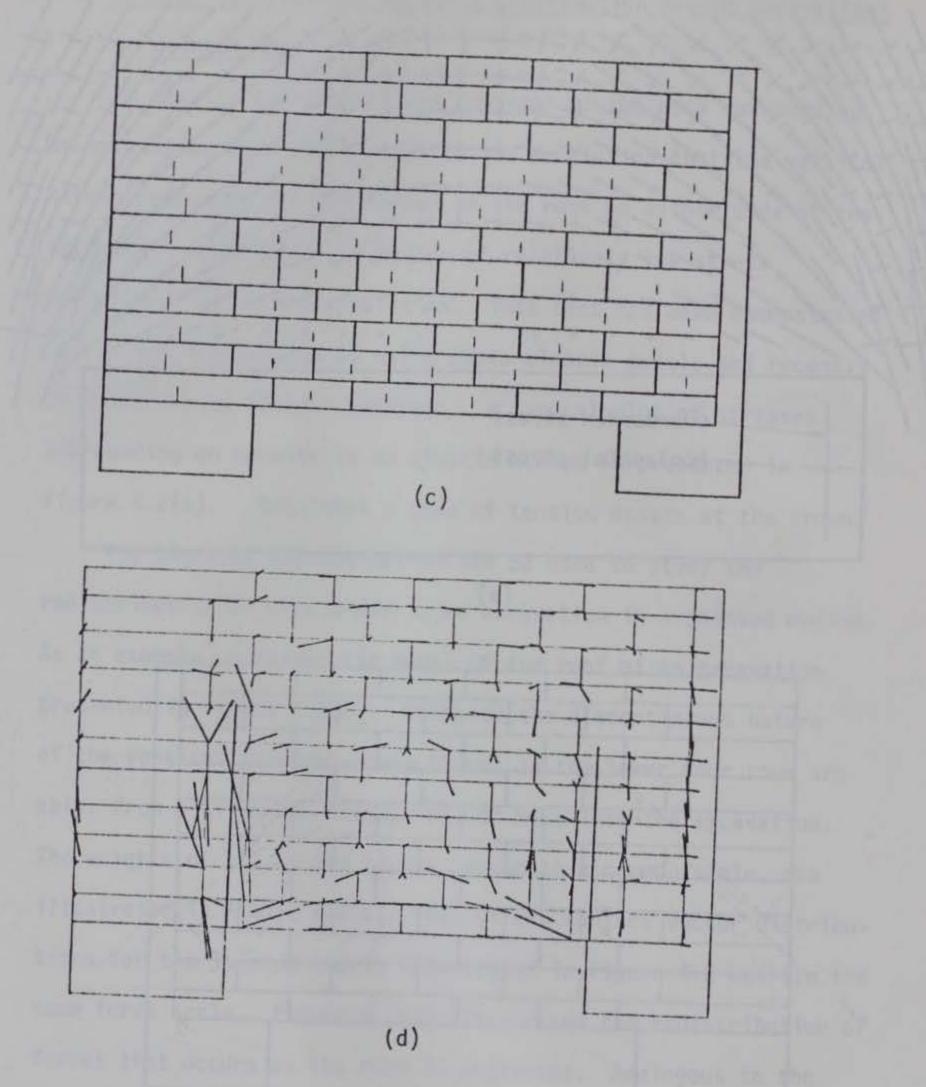


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Figure 4.2 (a) stress distribution in roof of opening in elastic medium; (b) model for behavior of jointed roof.

(b)

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Figure 4.2 (continued): (c) block weights for jointed roof model; (d) force distribution in roof following excavation (overburden due solely to block weight).

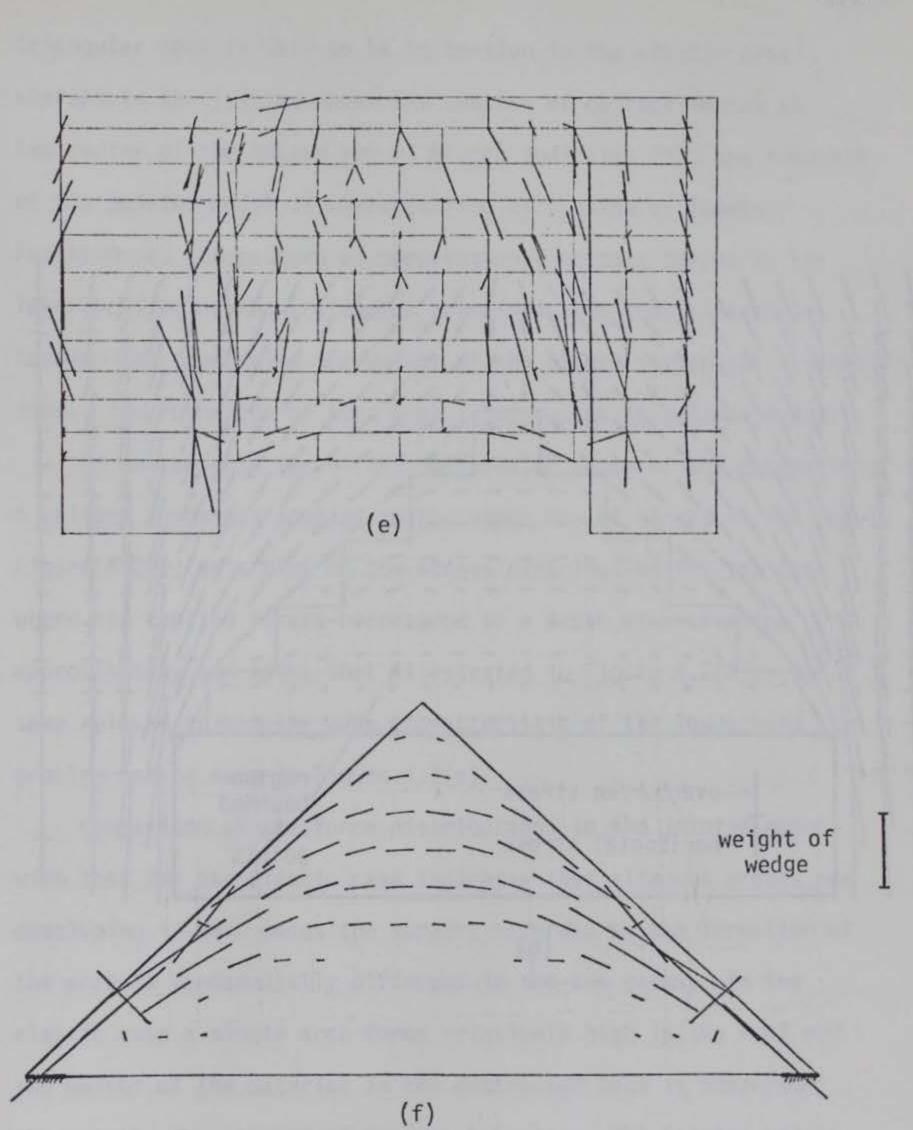


Figure 4.2 (continued: (e) force distribution in roof due to block weight and additional load to simulate greater depth: (f) stress distribution in triangular wedge supported at lower corners.

IV-10 11 region bounded by joints -overburden stress horizontal stress (g)

Figure 4.2 (continued): (g) stress distribution in jointed roof by Finite Element analysis.

triangular zone is seen to be in tension in the elastic case, whereas in the jointed model the absence of contact forces at the center of the bottom row of blocks indicates that the response of the jointed model is characterized by opening of joints. Furthermore, the pattern of compressional contact forces in the lower portion of the traingular zone indicates that an arch is forming and supporting the weight of the blocks within the triangular zone. The formation of this arch is discussed in section 4.3.3.

To investigate the effects of greater depth of the excavation, a uniform force was applied to the upper row of blocks in the model. Figure 4.2(e) is a plot of the stress distribution for the case where the applied forces correspond to a depth of excavation approximately ten times that illustrated in Figure 4.2(b). The same relaxed triangular zone characteristic of the low stress problem can be seen in Figure 4.2(e).

Comparison of the force distributions in the jointed models with that for the elastic case indicates that although arches are developing in both cases the support afforded by the formation of the arch is fundamentally different in the two cases. In the elastic case a single arch forms relatively high in the roof and the weight of the material in the destressed zone is supported through the development of tensional forces. The jointed models

on the other hand develop two arches, one relatively high in the roof which delineates the destressed zone; and one that acts to

support the lower strata.

This observation indicates a significant difference between the behavior predicted by elastic analyses and by the Distinct Element method. To determine to what extent the elastic behavior depended upon the continuity of the mass, several idealized models of roof behavior were analyzed, two of which are described here.

Figure 4.2(f) presents the results of a typical elastic analysis wherein the destressed zone was analyzed independently of the surrounding rock mass. The arch is still seen to form in the upper portion of the wedge of material and the material in the lower part of the wedge is in tension. This is in direct contrast to the behavior of the jointed masses analyzed by the Distinct Element method.

Figure 4.2(g) presents the results of a Finite Element analysis where the destressed zone was bounded approximately by a series of joint elements. Once again, the resultant behavior is characterized by a high arch and tensional forces; no evidence of arching action in the lower portion of the destressed zone is seen.

The behavior of the roof above an excavation in an elastic medium is thus seen to be fundamentally different than the behavior of a similar excavation in a jointed medium. The next portion of this chapter presents the results of an investigation to determine the causes of this fundamental difference.

4.3 A Model for the Behavior of Jointed Mine Roofs

The analyses discussed in this chapter deal with the behavior of the roofs of excavations in a medium where jointing is vertical and horizontal. The models have been kept simple deliberately so as to gain insight into relationships among the various parameters. As the overall goal of this study is to demonstrate the usefulness of the Distinct Element method in the analysis of excavation in jointed rock, more effort has been expended on demonstrating the effect of varying the significant parameters than on developing a single, all encompassing equation purported to describe the behavior of mine roofs.

The majority of the analyses to be discussed utilize similar jointed models, but although the chosen models are realistic the limitations were not imposed by the Distinct Element method as such; the techniques presented in this chapter are equally applicable to any model configuration. Although outside the scope of this study it is easy to envision an eventual compendium of various model geometries that portrays graphically the differences in the behavior of models.

4.3.1 The basic model

The basic model used for analysis consists of a rectangular

opening in a rock mass with continuous horizontal jointing and discontinuous jointing in the vertical direction as shown in Figure 4.3. This model does not consider the effect of joint inclination but does allow for variation of the span, aspect ratio of the blocks and friction angle of the joint surfaces.

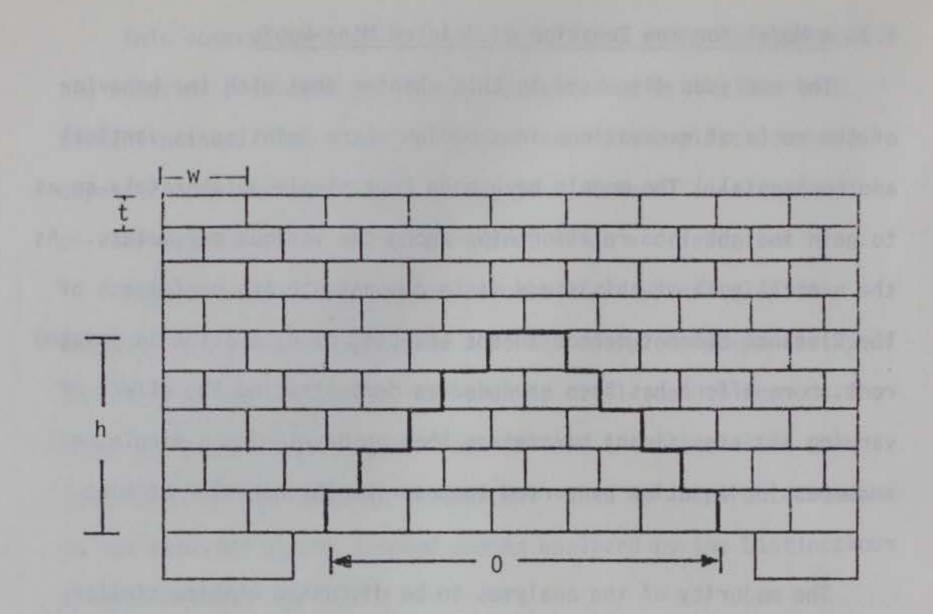
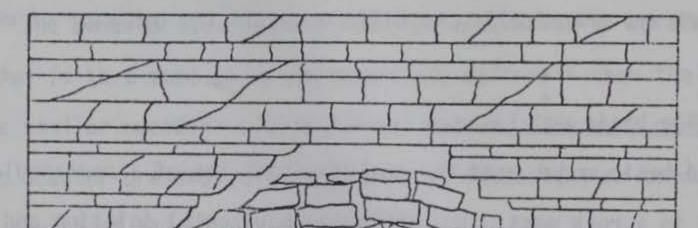


Figure 4.3 Jointed model upon which analysis was based. (O is span width, w is block width, t is block thickness and h is height of the triangular wedge.



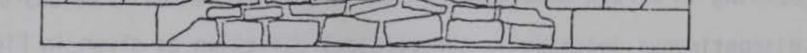


Figure 4.4 Diagramatic section of a roof fall (After Jones and Davies, 1929).

As justification for the use of the model a brief summary is given of four previous studies comprising theoretical calculations, laboratory as well as field observations and measurements, which utilized a similar model or support the model.

1) Behavior of Coal Mine Roofs

Jones and Davies (1929) presented a summary of their observations of roof behavior in British coal mines. They found that roof falls were invariably limited in height, the majority of the falls extending from 3 to 10 feet upward; falls exceeding 15 feet in height were considered exceptional. Judging from their description of the mining methods, the drifts were from 12 to 18 feet wide. They also concluded that the canopy of the fall was typically stepped along the sides "in the manner of a stairway viewed from below". A diagramatic section from their paper is reproduced in Figure 4.4.

2) Loads on Tunnel Supports

On the basis of observations and measurements of timber crushing in railway tunnels, Terzaghi (1946) proposed a classification scheme for the estimation of the maximum probable load on tunnel supports. Figure 4.5 presents one of the models used by Terzaghi to illustrate his concept that in relatively thin strata with many joints a peaked roof will develop. According to Terzaghi a constant

load with a height equal to the height of the peaked roof acts to load

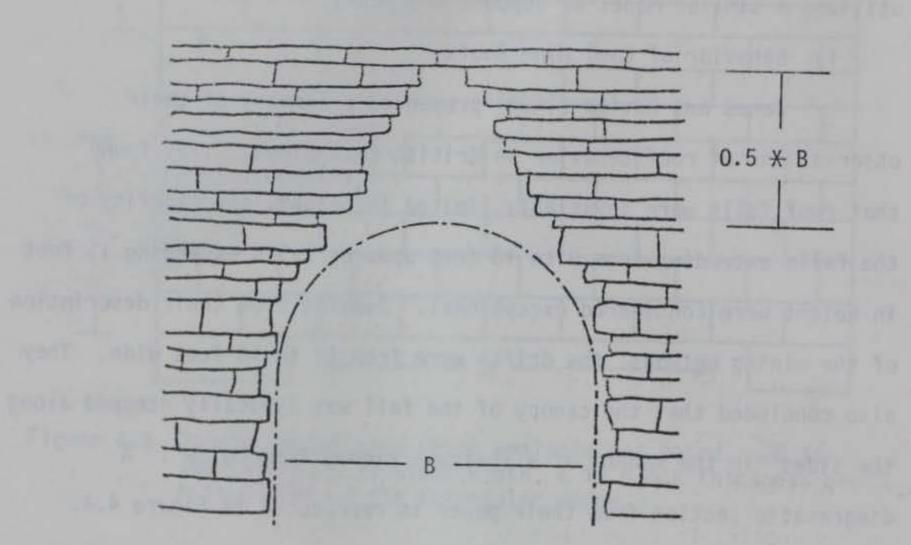
the tunnel supports.

3) Laboratory Investigation of Arching

Trollope (1966) utilized a physical model with continuous joints parallel to the roof and discontinuous jointing in the

BI-VI

perpendicular direction to demonstrate the behavior of an excavation roof. Like Terzaghi he concluded that in general, two zones may be identified within the immediate roof.



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Figure 4.5 Maximum probable overbreak if no support furnished (Terzaghi, 1946)

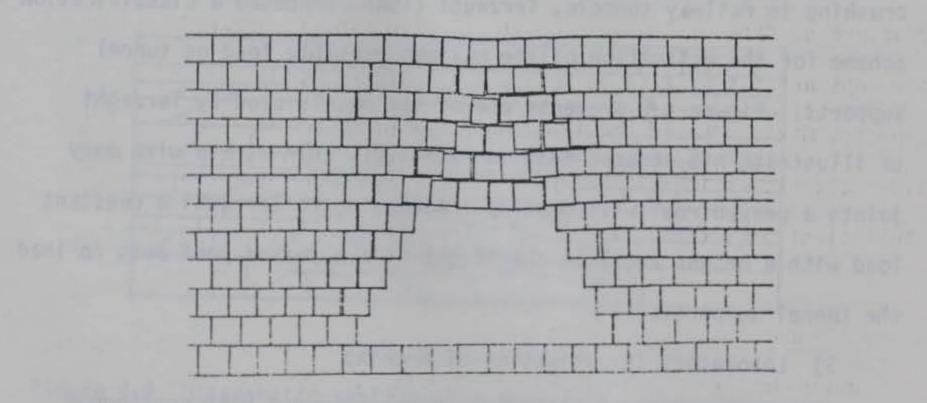


Figure 4.6 Trollope's Block Jointed Model (Trollope, 1966)

The first is inherently stable; the other zone which he referred to as the suspended zone, corresponds roughly with Terzaghi's triangular zone. Whereas Terzaghi concluded that the material within the zone would load the tunnel supports, Trollope was more concerned with the development of arching and stability within the suspended zone. Trollope's model is shown diagramatically in Figure 4.6.

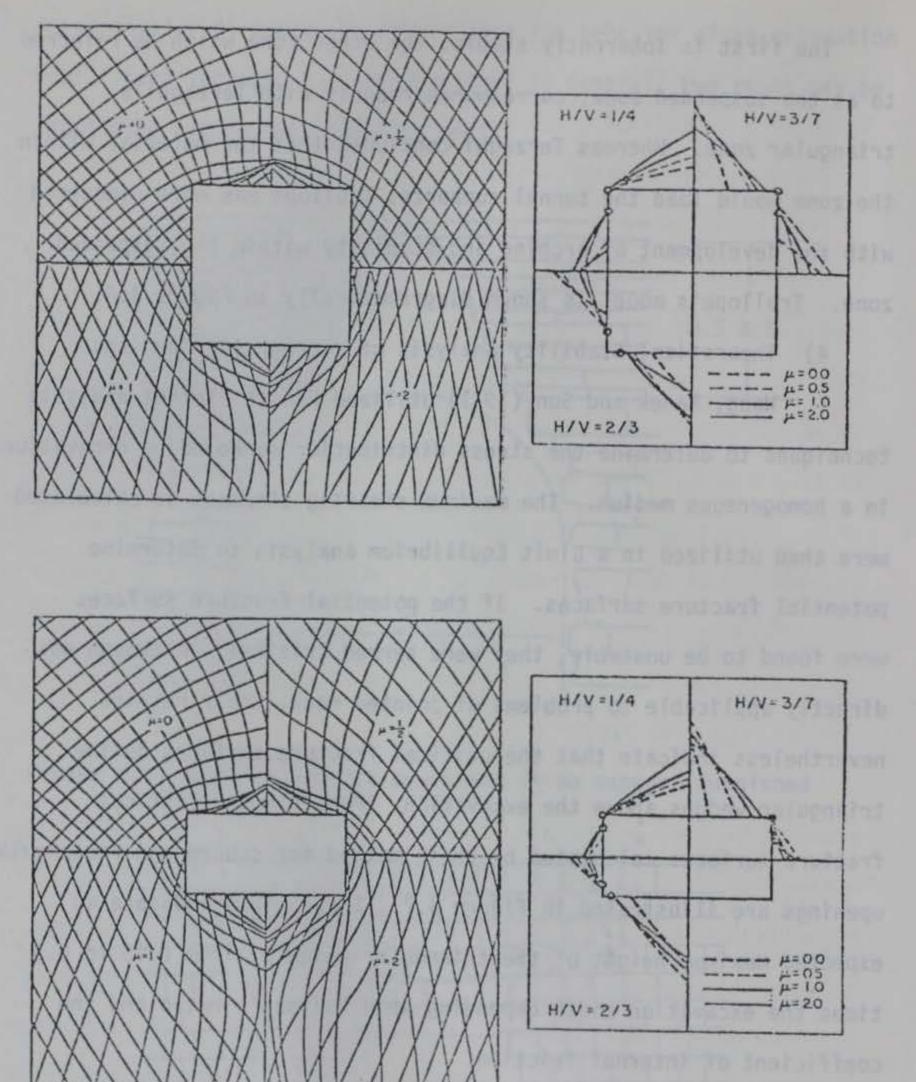
4) Theoretical Stability Analysis of Underground Openings

Wang, Panek and Sun (1971) utilized Finite Element analysis techniques to determine the stress distribution surrounding excavations in a homogeneous medium. The maximum shearing stresses so calculated were then utilized in a Limit Equilibrium analysis to determine potential fracture surfaces. If the potential fracture surfaces were found to be unstable, they were termed critical. Although not directly applicable to problems of jointed rock, their results nevertheless indicate that the critical fracture surfaces define triangular wedges above the excavation. Possible and critical fracture surfaces calculated by their method for square and rectangular openings are illustrated in Figure 4.7. These plots indicate an expected maximum height of the triangular wedge of from 0.15 to 0.5 times the excavation width depending upon Poisson's ratio and the coefficient of internal friction.

IV-17

4.3.2 Properties of the basic model

Referring once again to Figure 4.3 it can be seen that, by kinematic considerations, a triangular wedge of material is free to



H/V = ratio of horizontal to vertical stress μ = coefficient of internal friction

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Figure 4.7 Possible and critical fracture surfaces for square and rectangular openings. (Wang, Panek and Sun, 1971)

move into the excavation. The height of this triangular wedge (referred to by Terzaghi as overbreak and by Trollope as the height of the suspended zone) is easily calculated in terms of the excavation span and the thickness and width of the blocks defined by the jointing pattern.

The number of blocks (b) in the bottom row of the roof strata is given by:

b = 0/w

O is the true span of the excavation

w is the block width

(Note that span is defined as illustrated in Figure 4.3) Restricting the analyses to the case where all blocks are identical, it is easily verified that the height of the triangular wedge is given by:

where: t is the block thickness

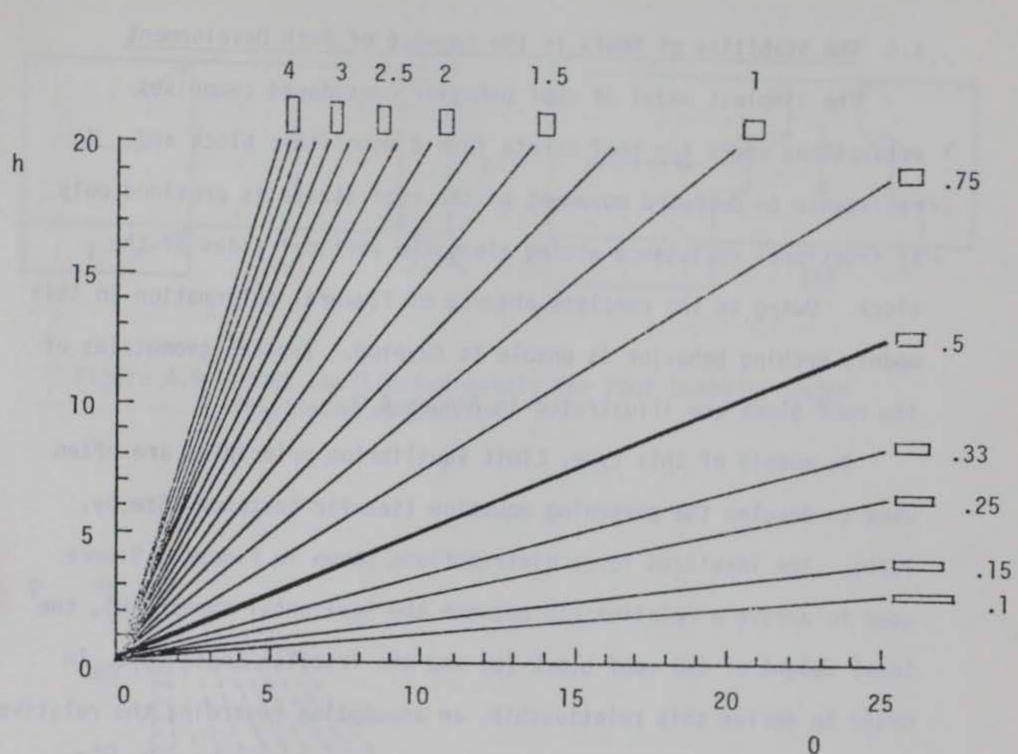
In terms of the aspect ratio of the blocks (A = t/w)

h = 0 . A 4.2

Equation 4.2 is plotted in Figure 4.8 as a family of curves representing the wedge height as a function of span for various aspect ratios; the block shapes are also illustrated for several values of the aspect ratio. The curves represent kinematic considerations only and indicate that increasing the aspect ratio of the blocks has the effect of increasing the height of the traingular wedge and thus, for a constant block width, the volume of material that tends to move into the excavation. The curve corresponding to an aspect ratio of 0.5 is plotted more boldly since this is the equation for the height of the arch in stratified rock according to Terzaghi.

The graph is presented without units since the axes are consistent; that is, if the span is measured in meters, then the height of the wedge will be in meters.

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Figure 4.8 Relationship between span width (0), and height of suspended zone (h) for various values of the aspect ratio (t/w) of the model illustrated in Figure 4.3. The aspect ratio of the blocks is graphically portrayed.

4.4 The Stability of Roofs in the Absence of Arch Development

The simplest model of roof behavior considered comprises excavations where the roof strata form a monolithic block and resistance to downward movement of the roof strata is provided only by frictional resistance acting along the vertical sides of the block. Owing to the complete absence of flexural deformation in this model, arching behavior is unable to develop. Typical geometries of the roof block are illustrated in Figure 4.9.

In models of this type, Limit Equilibrium principles are often used to develop the governing equation (see for instance, Szechy, 1970). The idealized force distributions shown in Figure 4.9 were used to derive a relationship between the horizontal thrust (H), the total weight of the roof block (W) and the friction angle (ϕ). In order to derive this relationship, an assumption regarding the relative magnitudes of the frictional reaction (R1, etc.) must be made. To make the models illustrated in Figure 4.9 statically determinate two assumptions must be made: first, it is assumed that full frictional resistance is mobilized at all points of contact; and, second, it is assumed that the frictional resistance vectors are symmetric about the block. Under these assumptions, equilibrium principles can be used to derive the equation relating horizontal force to block weight and friction angle. This relationship is: $H = 1/2 W \cot \phi$ 4.3

A number of monolithic roof geometries were analyzed by the Distinct Element method for purposes of comparison to equation 4.3. The results of these analyses are presented in Figure 4.10 where the joint plane angle of friction required for stability is plotted as out of the quality ber 100 that Thete I had the root asigned. The

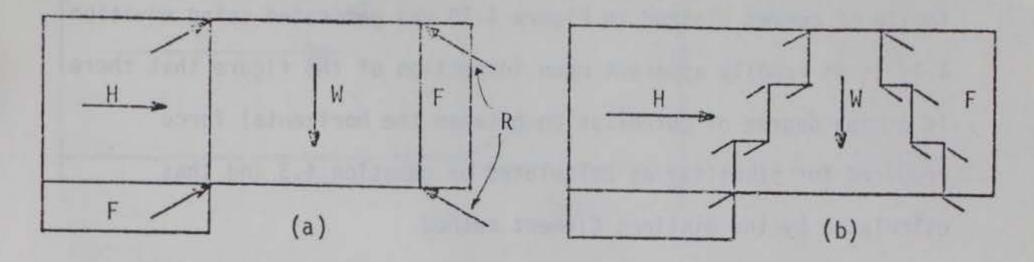
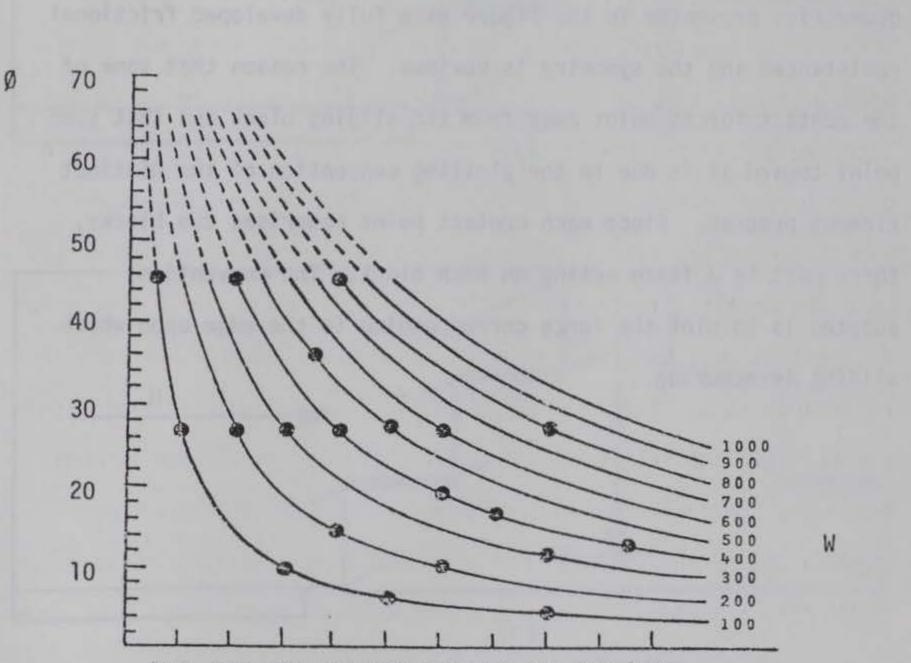


Figure 4.9 Limit Equilibrium models for roof behavior under frictional suspension.



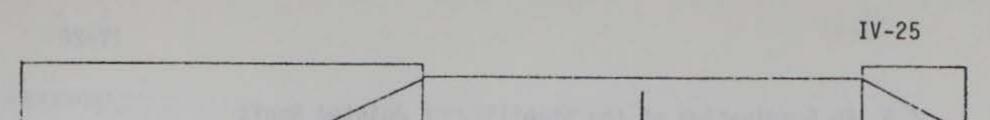
100 200 300 400 500 600 700 800 900 1000

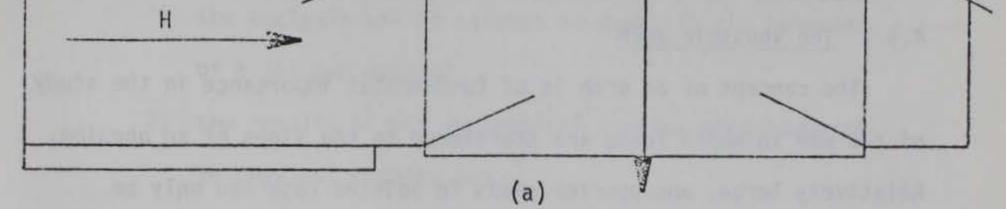
Figure 4.10 Friction angle (Ø) required for stability as a function of horizontal force (H) and roof weight (W) in a non arching model.

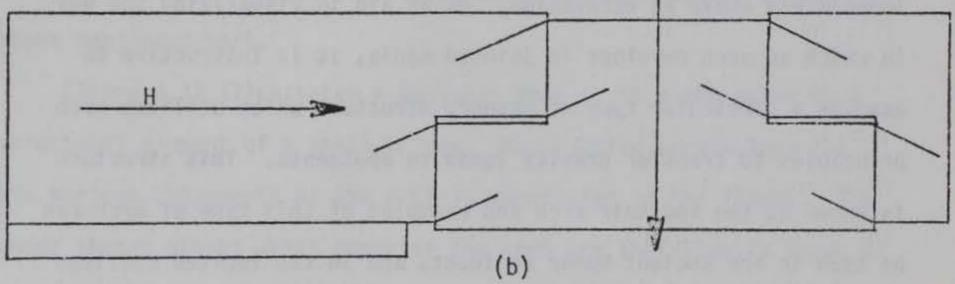
H

a function of the applied horizontal force and the roof weight. The family of curves plotted in Figure 4.10 was generated using equation 4.3; it is readily apparent upon inspection of the figure that there is a high degree of correllation between the horizontal force required for stability as calculated by equation 4.3 and that calculated by the Distinct Element method.

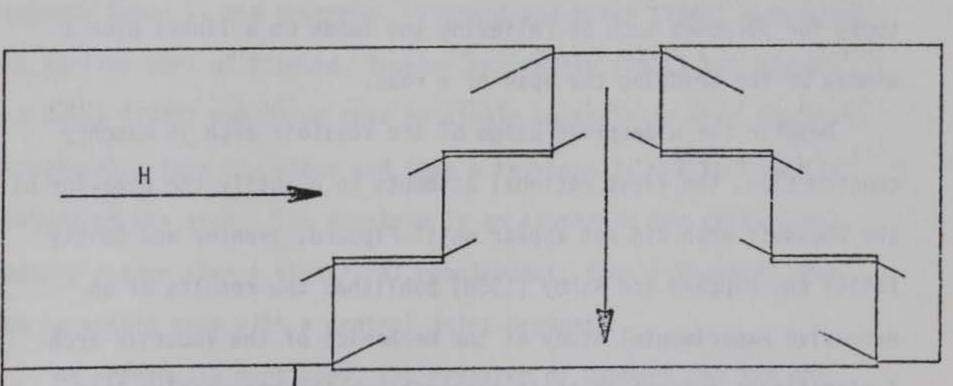
In the derivation of equation 4.3 it was assumed that full frictional resistance was developed at sliding contacts and that the frictional resistance developed symmetrically. Figure 4.11 illustrates that this is indeed the case; the three representative geometries presented in the figure have fully developed frictional resistances and the symmetry is obvious. The reason that some of the contact forces point away from the sliding block and that some point toward it is due to the plotting convention of the Distinct Element program. Since each contact point comprises two blocks, there must be a force acting on each block. The convention adopted is to plot the force corresponding to the edge upon which sliding is occuring.







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(c)

Figure 4.11 Frictional resistance developed in no-arching models at onset of sliding failure.

4.5 An Examination of the Stability of Jointed Roofs

4.5.1 The Voussoir arch

The concept of an arch is of fundamental importance in the study of the way in which loads are transfered to the sides of an opening. Relatively large, unsupported spans in jointed rock can only be obtained if the major portion of the load due to the overlying strata is carried to the abutments through arches forming in the jointed rock immediately above an excavation. As an aid in visualizing the way in which an arch develops in jointed media, it is instructive to examine a particular type of masonry structure which utilizes arch principles to transfer gravity loads to abutments. This structure is known as the Voussoir arch and examples of this type of arch can be seen in the ancient Roman aquiducts and in the vaulted ceilings of European cathedrals. The Voussoir arch is still in common use today for purposes such as relieving the loads on a lintel over a window or for bridging the span of a road.

Despite the widespread usage of the Voussoir arch in masonry construction, the first rational attempts to quantify the behavior of the Voussoir arch did not appear until Pippard, Tranter and Chitty (1936) and Pippard and Ashby (1938) published the results of an extensive experimental study of the mechanics of the Voussoir arch.

A significant outcome of their research was the observation that a Voussoir arch could be analyzed as a three hinged, and thus statically determinate, arch.

The analyses performed by Pippard, Tranter and Chitty and Pippard and Ashby are significant to this present study for at least three

reasons:

- the analysis was an attempt to quantify the behavior of a jointed medium;
- the results of the theoretical studies were compared to physical models; and
- 3) the method of analysis introduces the general

calculation techniques of linear arch analysis. It would seem worthwhile, therefore, to devote some detail to the above mentioned work.

Figure 4.12 illustrates a Voussoir arch as it might occur as a structural element of a small bridge. Descriptive terminology for the various components of the arch is identified in the figure. The wedge shaped blocks which comprise the arch are individually known as voussoirs; they are usually disposed symmetrically about a central voussoir known as the keystone. Pippard and Baker (1948) summarized the earlier work of Pippard, Tranter and Chitty (1936) and Pippard and Ashby (1938) and noted that no single voussoir is more important structurally than any other and that a keystone is not an essential feature of the arch. The keystone is an aesthetic and traditional feature rather than a structural requirement; thus a Voussoir arch can be stable even with a central joint present.

As previously mentioned, the research of Pippard and his coworkers indicated that the force distribution in a Voussoir arch would be statically determinate, in the absence of fixity at the abutments, owing to the development of three hinges. For a symmetrically loaded Voussoir arch two of the hinges were seen to be loacted at the

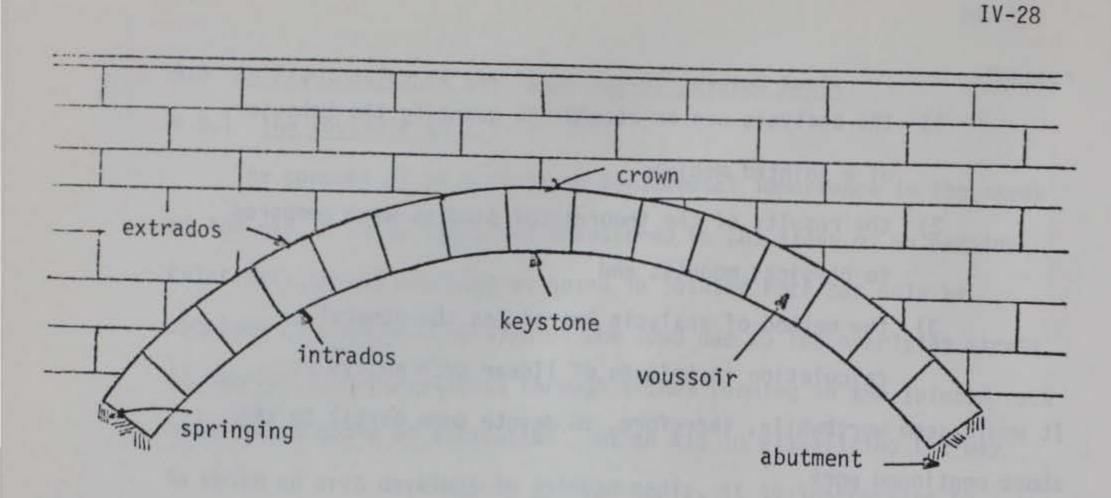


Figure 4.12 A typical Voussoir arch application with component parts identified.

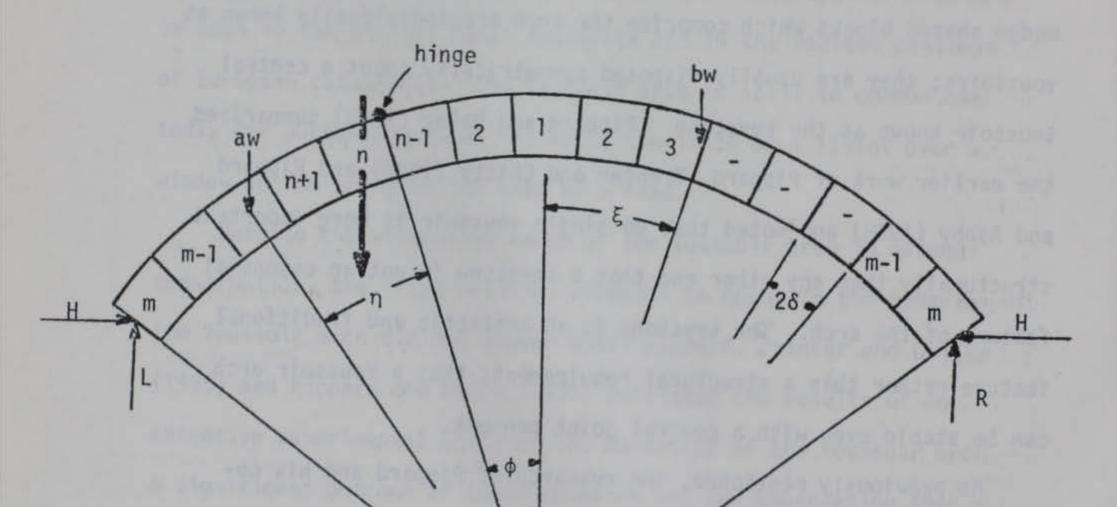


Figure 4.13 Nomenclature used in analysis of a non-symmetrically loaded Voussoir arch. For a description of identified variables see the text.

θ.

abutments with the third hinge at midspan if a central joint existed or on one of the faces of the keystone if it were present. For the case of non-symmetrical point loading the two abutment hinges developed as in the symmetrical case, but the position of the third hinge was initially variable, typically located somewhere on the extrados between midspan and the loaded voussoir. Increased load or abutment movement caused the position of the variable hinge to move closer to the loaded voussoir; when the hinge reached the joint next to the loaded voussoir on the midspan side, it did not change its position again until failure had occured.

The observations concerning the formation of hinges, coupled with the results of the other analytical and experimental studies performed by Pippard and his co-workers provide good data for checking the accuracy of the Distinct Element method as well as introducing the techniques of linear arch analysis which will be used extensively in this chapter.

The idealized model used in the present study is illustrated in Figure 4.13. The model arch is circular in shape and the abutments subtend an angle of 20. Hinges are assumed to develop at the abutments and at the extrados of the joint nearest the point of application of the external load W on the side nearest the crown.

Each individual voussoir subtends an angle of 2δ and has a weight w. The voussoirs are numbered consecutively from 1 at the keystone to m at the abutment; thus the total number of voussoirs in the arch is 2m-1. In addition to the external load, the arch is also loaded by its self weight. With respect to the non-abutment hinge, self weights of magnitude aw and bw act on the shorter and longer spans respectively, as illustrated in Figure 4.13. The points of application of the loads are located as follows: the external load W is applied at the centroid of voussoir number n; the longer span load is located at an angle ξ clockwise from the vertical; the shorter span load is located at an angle n counter clockwise from the hinge which in turn is located at an angle ϕ counter clockwise from the vertical. It is easily shown that for an odd number of voussoirs;

> $n = \xi = (m - n + 1) \delta;$ $\phi = (2n - 3) \delta;$ $\theta = (2m - 1) \delta;$ a = m - n + 1; andb = m + n - 2

For a Voussoir arch with an even number of voussoirs a slight modification must be introduced; the voussouirs are numbered consecutively from the crown joint starting with 1 and ending with m. Thus, these are 2m voussoirs in the arch. The corresponding parameters are given by:

> $n = \xi = (m - n + 1) \delta;$ $\phi = 2(n - 1) \delta;$ $\theta = 2 m \delta;$

4.4b

a = m - n + 1; and b = m + n - 2 The analytical approach used by Pippard, Tranter and Chitty (1937) involved the determination of strain energies and application of Castigliano's theorems. This approach was necessary because they were interested in displacements as well as forces and because they analyzed indeterminate as well as determinate arches. Since the present study is limited to three hinged arches which are statically determinate, a simpler analytical method has been adopted. Equilibrium principles provide the means to determine the force distribution in a statically determinate structure and have been used to derive the following equations.

The horizontal force H induced by a point load of magnitude W applied at the centroid of voussoir n subject to the development of hinges in the manner previously described is found by the superposition of the horizontal force H_W due to the external load and the horizontal force H_S due to the self load. These horizontal forces are calculated by taking moments about the midspan hinge and using an equation expressing vertical equilibrium.

The horizontal thrust due to the self weight of the arch is given by:

$$H_{s} = ((\sin\theta - \sin\phi) L_{s} - aw (\sin (\phi + \eta) - \sin\phi)) \frac{1}{\cos\phi - \cos\theta}$$
 4.5

The quantity L_s represents the vertical abutment reaction on the shorter span due to the self weight of the arch and is given by:

$$L_{c} = ((\sin\phi + \sin(\theta + n)) aw + (\sin\theta - \sin n) bw) \frac{1}{2 \sin\theta}$$
 4.6

The horizontal thrust due the applied point load is given by:

 $H_{W} = \left(L_{W} (\sin\theta - \sin(\phi + \delta) - W(\sin(\phi + \delta) - \sin\phi)\right) \frac{1}{\cos\phi - \cos\theta} \quad 4.7$

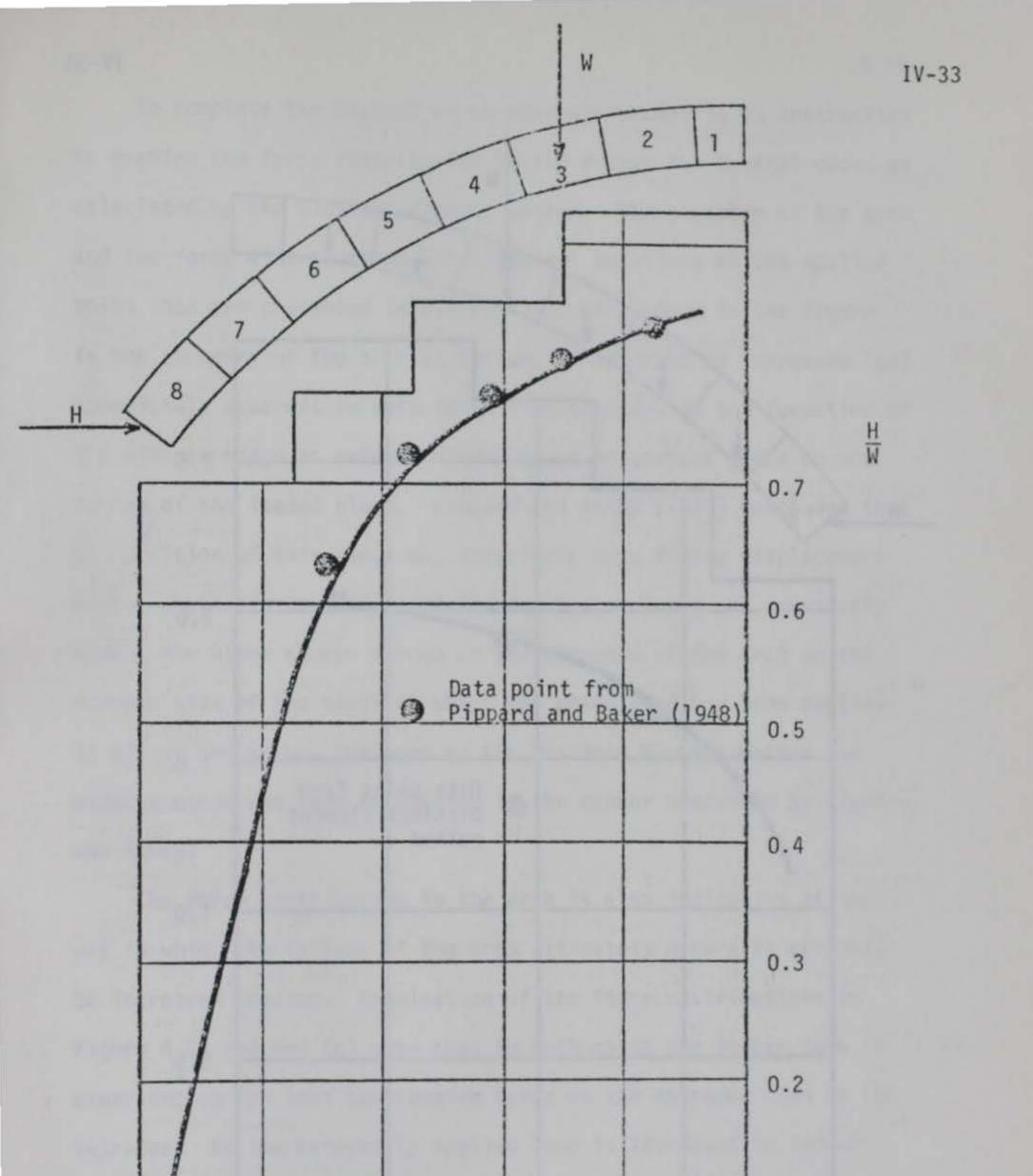
The quantity L_W represents the vertical abutment reaction on the shorter span due to the point load and is given by:

$$L_{W} = \frac{W}{2} \left(1 + \frac{\sin(\phi + \delta)}{\sin \theta} \right)$$
4.8

To demonstrate the validity of the above equations, several data points from Pippard and Baker (1948) are plotted in Figure 4.14a with the plotted curve representing the ratio of horizontal force to applied load, neglecting the self weight of the arch, given by equations 4.7 and 4.8. Since Pippard and Baker did not present their analytical expressions for the ratio of horizontal thrust to applied load, the parameters used in equations 4.7 and 4.8 were scaled from drawings in their paper. In light of this limitation, the fit of the data points to the theoretical expression can be described as quite good.

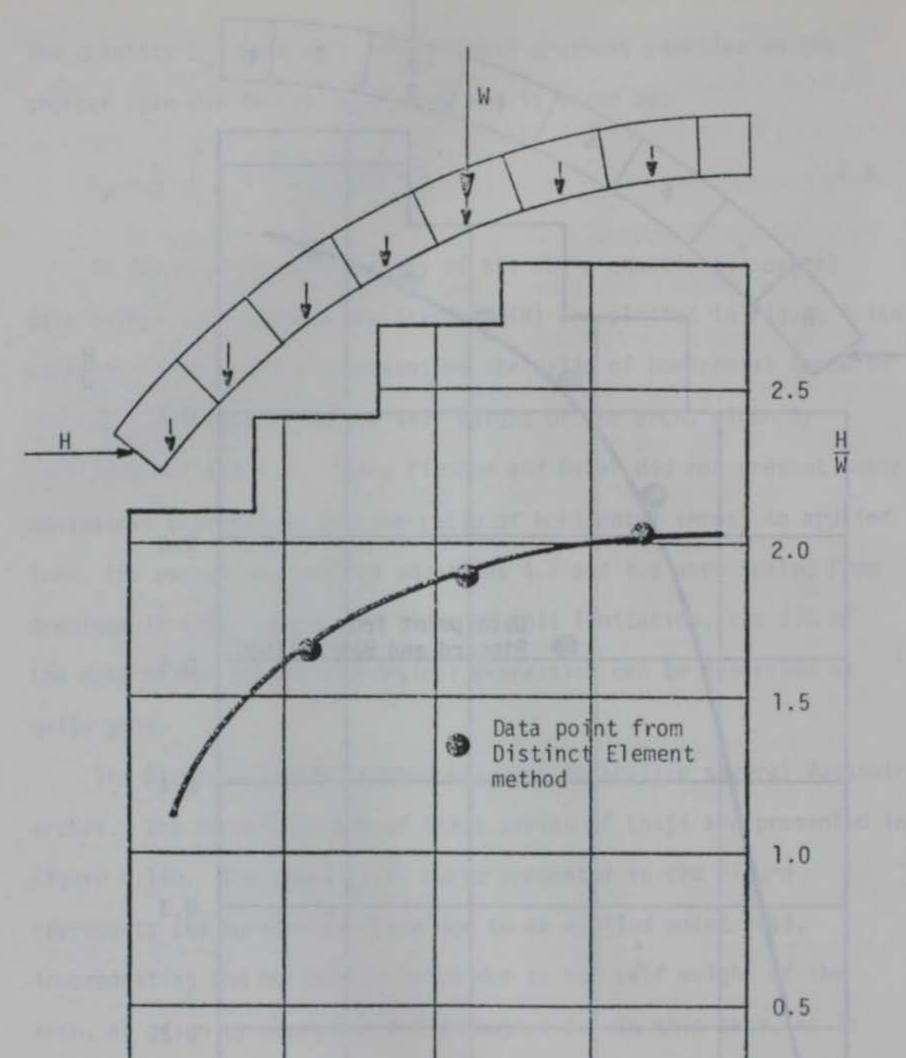
The Distinct Element method was used to analyze several Voussoir arches. The results of one of these series of tests are presented in Figure 4.14b. The theoretical curve presented in the figure represents the horizontal force due to an applied point load, incorporating the horizontal force due to the self weight of the arch, as given by equations 4.5 through 4.8. In this case, as in other Voussoir arches analyzed by the Distinct Element method, the

test points fit the theoretical curve quite well, and suggest that the Distinct Element method is capable of reproducing the results of the physical model tests performed by Pippard and his co-workers.



0.1

Figure 4.14(a) Horizontal thrust developed due to an applied point load neglecting the self weight of the arch.



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Figure 4.14(b) Horizontal thrust due to an applied point load incorporating the self weight of the arch.

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To complete the discussion on Voussoir arches it is instructive to examine the force distribution in the arches for several cases as calculated by the Distinct Element method. The geometry of the arch and two force distributions for different positions of the applied point load are presented in Figure 4.15; also shown in the figure is the geometry of the arch at failure in response to increased load. Immediately apparent in both force distributions is the formation of the midspan hinge as evidenced by absence of contact force on one corner of the loaded block. Pippard and Ashby (1938) concluded that the position of this hinge was invariable once finite displacement of the abutments or sufficient loading had occured. As previously noted, the hinge always formed on the extrados of the arch on the midspan side of the block to which the point load had been applied; in all of the arches analyzed by the Distinct Element method the midspan hinge was seen to develop in the manner described by Pippard and Ashby.

The force distribution in the arch is also indicative of the way in which the failure of the arch ultimately occurs in response to increased loading. Examination of the force distributions in Figure 4.15 (b) and (c) show that in both cases the longer span is experiencing far less compressive force on the extrados than on the intrados. As the externally applied load is increased to induce failure, the geometry shown in Figure 4.15(d) develops. The increased load leads to the development of a fourth hinge on the arch at which point the arch collapses. The position of the fourth

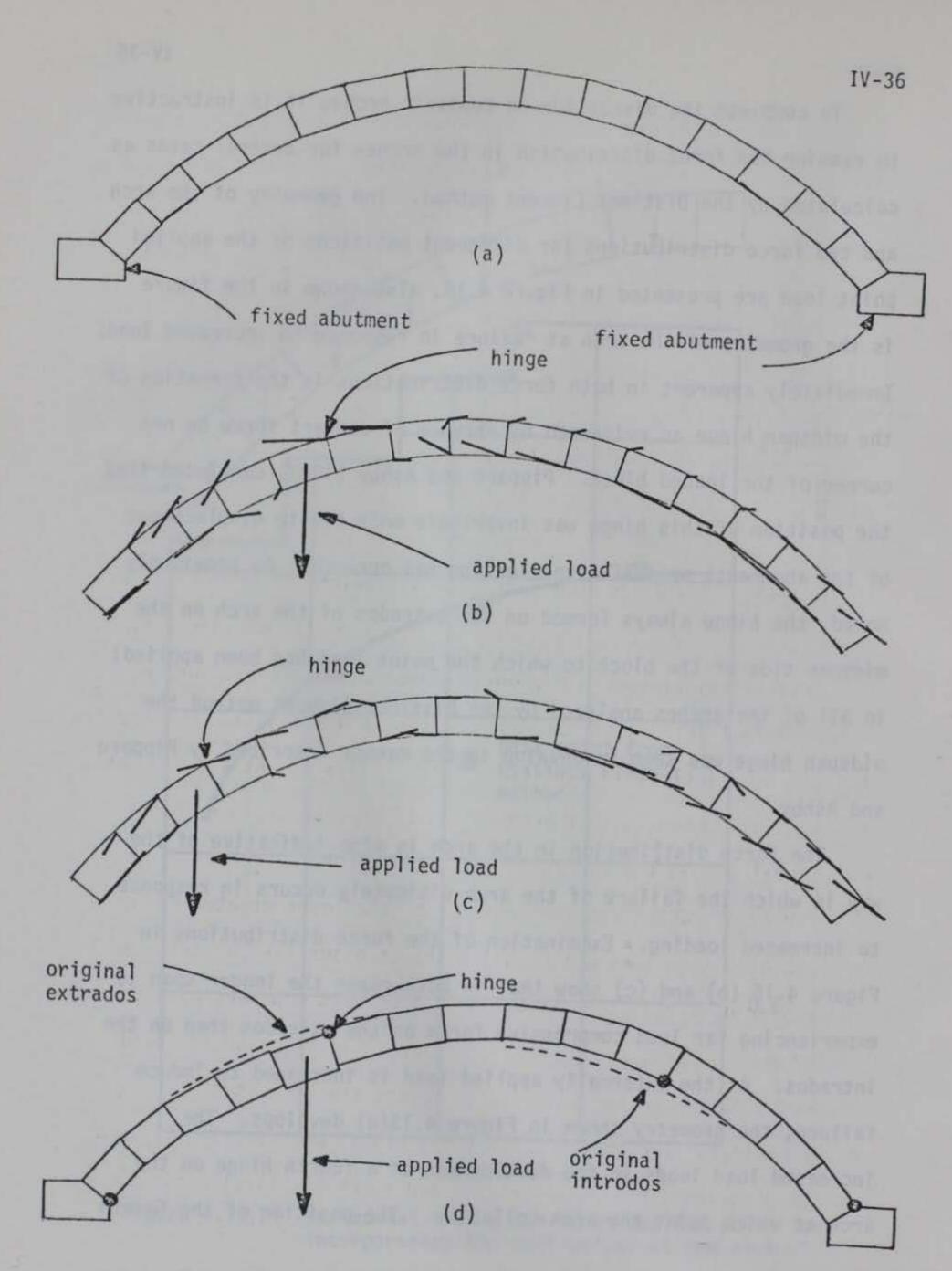


Figure 4.15 Variation in force distribution with the position of the applied load, and the ultimate collapse of a Voussoir Arch.

hinge is not as predictable as the other three, and is complicated by the fact that slippage may occur along the sides of the voussoirs. The method of calculation of the critical external load, which involves trial and error procedures and is beyond the scope of this brief introduction to Voussoir arches, is discussed by Pippard and Baker (1948).

4.5.2 Arching conditions in jointed roofs

As early as 1885 (Jones and Davies, 1929) Fayol demonstrated that an arching action could occur in bedded roofs and would act to shield the immediate roof from the full weight of the overlaying material. The fact that the height of the dome formed when a mine roof failed was limited was taken by Jones and Davies as further evidence that arching action was occurring and acting to transfer the bulk of the vertical load to the adjacent pillars. At a later date, Evans (1941) proposed that arching was also occurring within the immediate roof in the manner of a Voussoir Arch.

Evans characterized the behavior of the lower strata in a mine roof as a jointed beam within which the stresses were distributed in the manner of a modified three hinged arch. As downward displacement of the beam occurs, the central joint opens in response to "bending"

induced tension and the compressive forces are increased at the upper contact. The analogy to a three hinged arch is clearly seen in the postulated pressure distribution which is illustrated in Figure 4.1. Because the manner in which the forces are distributed resembles the classical Voussoir arch, this type of analysis is often referred to as Voussoir beam analysis.

Evans' research, and that which followed, was concerned with the stress state and subsequent fracture of the strata within the immediate roof above the excavation and is not directly applicable to the present study. The concept of two separate pressure arches in the roof strata is, however, of interest.

In the discussions that follow, the pressure arch that carries the weight of the superincumbent strata to the sides of the excavation will be termed the ground arch; the lower arch that forms within the wedge of failing material will be termed the roof arch.

The analyses that form the basis for the discussion presented in this chapter indicate clearly that the stability of the roof of an excavation in jointed material is dependent upon the formation of the roof arch. In fact, the general pattern of force distribution in the basic model of this study is that illustrated in Figure 4.2(d). Most of the weight due to the overlaying strata is transferred to the abutments through the ground arch; the stability of the resulting destressed zone is maintained through the development of the roof arch in the lower strata. Specific departures from this general pattern were observed in those

instances where the horizontal stress field was greater than that required for stability and in those instances where the block thicknesses exceeded some critical thickness. Both of these occurrences inhibit block rotations and thus the development of arching.

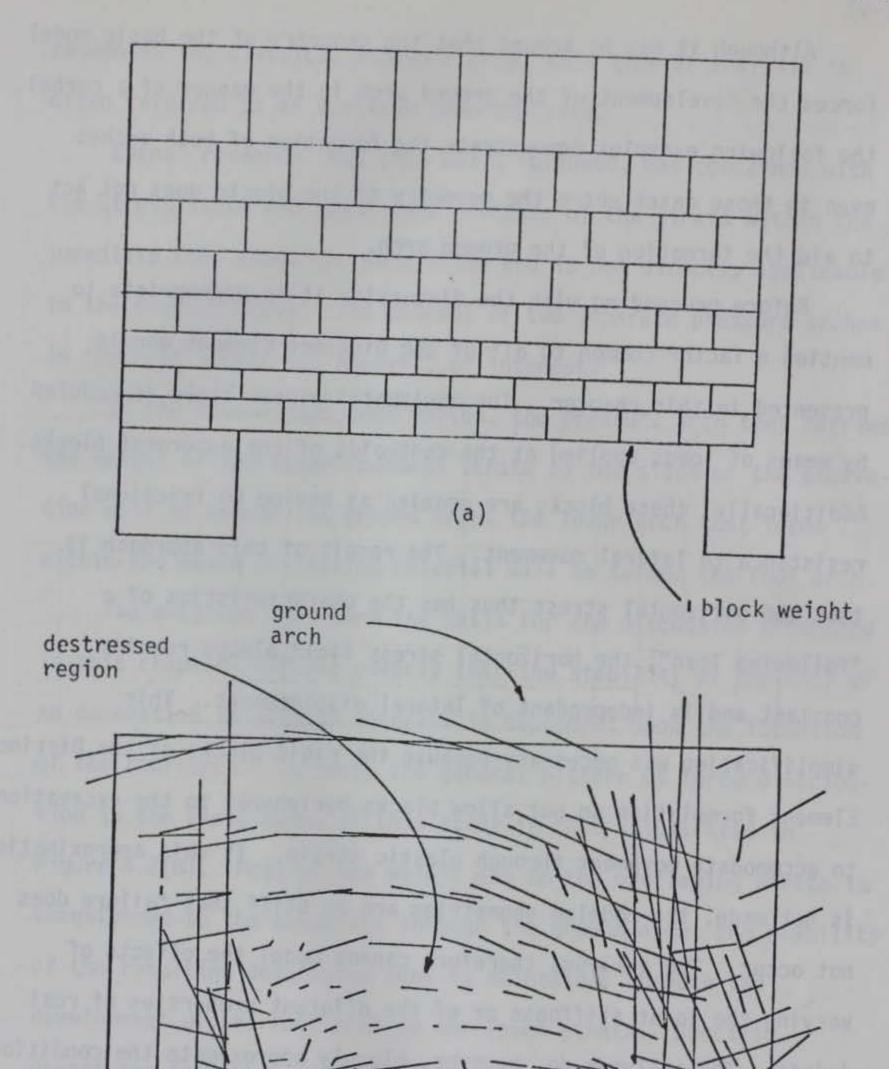
Although it may be argued that the geometry of the basic model forces the development of the ground arch in the manner of a corbel, the following examples demonstrate the formation of both arches even in those cases where the geometry of the blocks does not act to aid the formation of the ground arch.

Before proceeding with the discussion it is appropriate to mention a factor common to all of the Distinct Element models presented in this chapter. The horizontal stress field is modeled by means of loads applied at the centroids of the outermost blocks. Additionally, these blocks are modeled as having no frictional resistance to lateral movement. The result of this approach is that the horizontal stress thus has the characteristics of a "following load"; the horizontal stress field always remains constant and is independent of lateral displacement. This simplification was necessary because the rigid blocks of the Distinct Element formulation do not allow blocks peripheral to the excavation to accomodate movement through elastic strain. If this approximation is not made, the modeled geometries are so stiff that failure does not occur. The analyses therefore cannot model the effects of varying the joint stiffness or of the dilatant properties of real joints. The analyses do, however, closely approximate the conditions modeled by linear arch analysis and are considered to be valid,

though rudimentary, approaches to modeling the behavior of excavation roofs. Figure 4.16(a) illustrates an example of the basic model; if

complete failure were to take place, blocks from the lower six





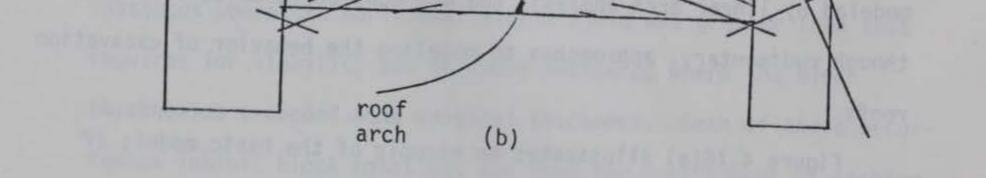
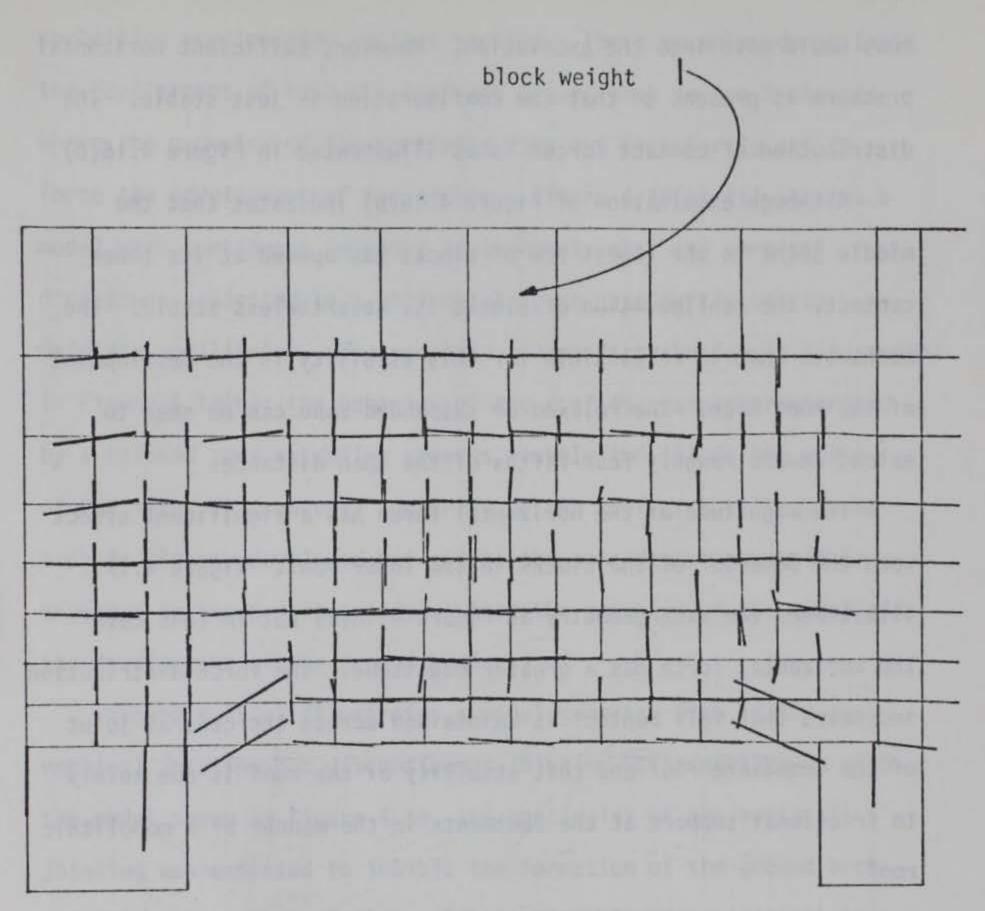


Figure 4.16 Formation of the ground and roof arches in a vertically discontinuous jointed model.





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Figure 4.17 Roof and ground arch development inhibited due to high horizontal forces.

rows would move into the excavation. However, sufficient horizontal pressure is present so that the configuration is just stable. The distribution of contact forces is as illustrated in Figure 4.16(b).

Although examination of Figure 4.16(b) indicates that the middle joint in the lowest row of blocks has opened at its lower contact, the configuration of blocks is, nevertheless stable. The mechanism that is responsible for this stability is the development of the roof arch. The relaxed or suspended zone can be seen to extend upward roughly four-fifths of the span distance.

The magnitude of the horizontal force has a significant effect upon the behavior of the blocks in the lower roof. Figure 4.17 illustrates the same geometry as Figure 4.16(a) but in this case the horizontal force has a greater magnitude. The force distribution indicates that full contact is maintained across the central joint of the immediate roof and that stability of the roof is due solely to frictional support at the abutments in the manner of a monolithic roof.

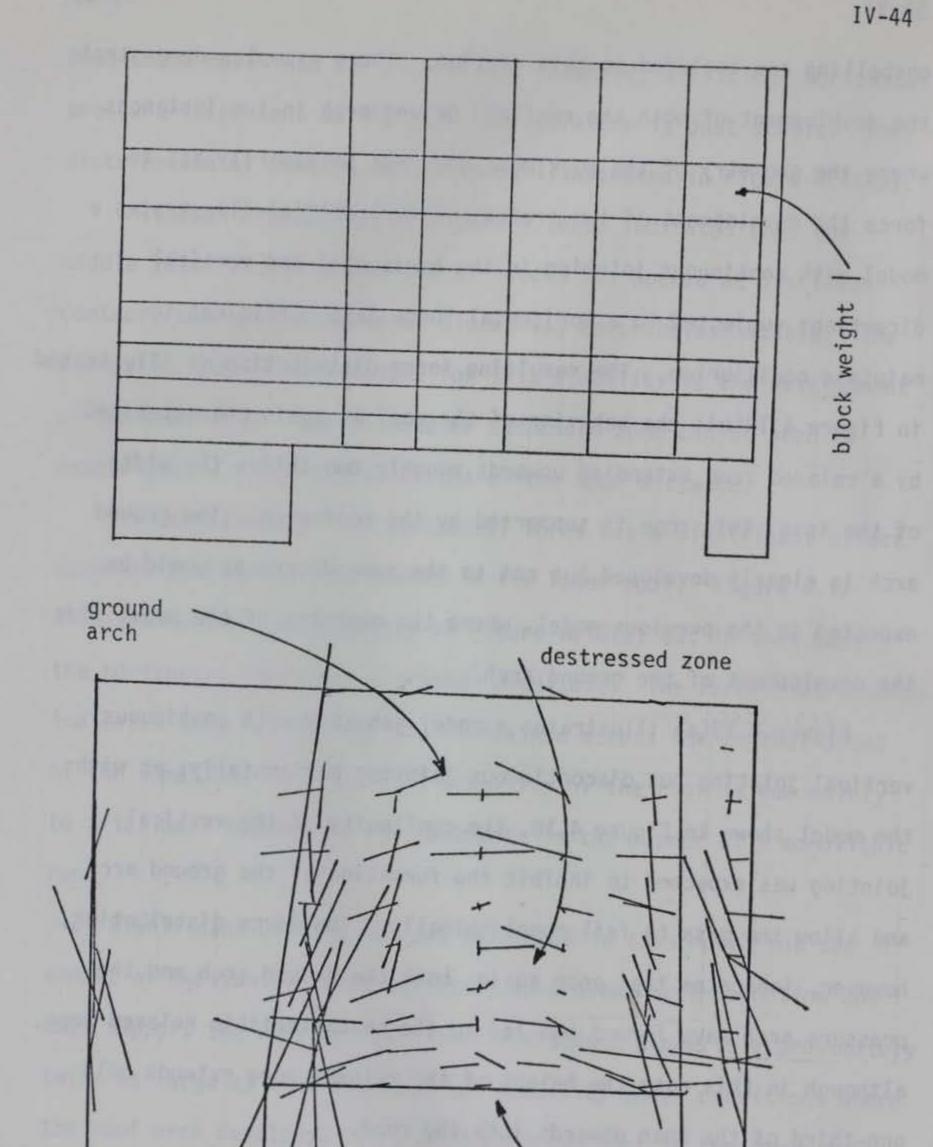
Significant arching has not developed in this model but the amount of horizontal force necessary to prevent arch formation and thus support the roof by frictional resistance alone is approximately twice as large as that required for stability under conditions where the roof arch develops. It should be noted that if the lower roof comprised a single block, the amount of force required to stabilize the configuration by frictional resistance would be less than the case where arching develops.

Two examples where the jointing pattern does not involve

corbelling are included in this section. These examples demonstrate the development of both the roof and ground arch in two instances where the geometry of the rock mass does not necessarily act to force the development of two arches. Figure 4.18(a) illustrates a model with continuous jointing in the horizontal and vertical directions subjected to a horizontal force just sufficient to maintain equilibrium. The resulting force distribution is illustrated in Figure 4.18(b); the behavior of the roof is again characterized by a relaxed zone extending upwards roughly two-thirds the width of the span. This zone is supported by the roof arch. The ground arch is clearly developed but not to the same degree as would be expected in the previous model, where the geometry of the model aids the development of the ground arch.

Figure 4.19(a) illustrates a model geometry with continuous vertical jointing but discontinuous jointing horizontally; as with the model shown in Figure 4.18, the continuity of the vertical jointing was expected to inhibit the formation of the ground arch and allow the mass to fail monolithically. The force distribution, however, indicates that once again, both the ground arch and the pressure arch have formed and led to the characteristic relaxed zone, although in this case the height of the relaxed zone extends only one-third of the span upwards into the roof.

The block movements that lead to the development of arches are primarily of a rotational nature. The rotations arise as the unequal forces on opposite sides of a block, which arise as the blocks move,





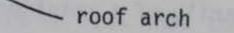
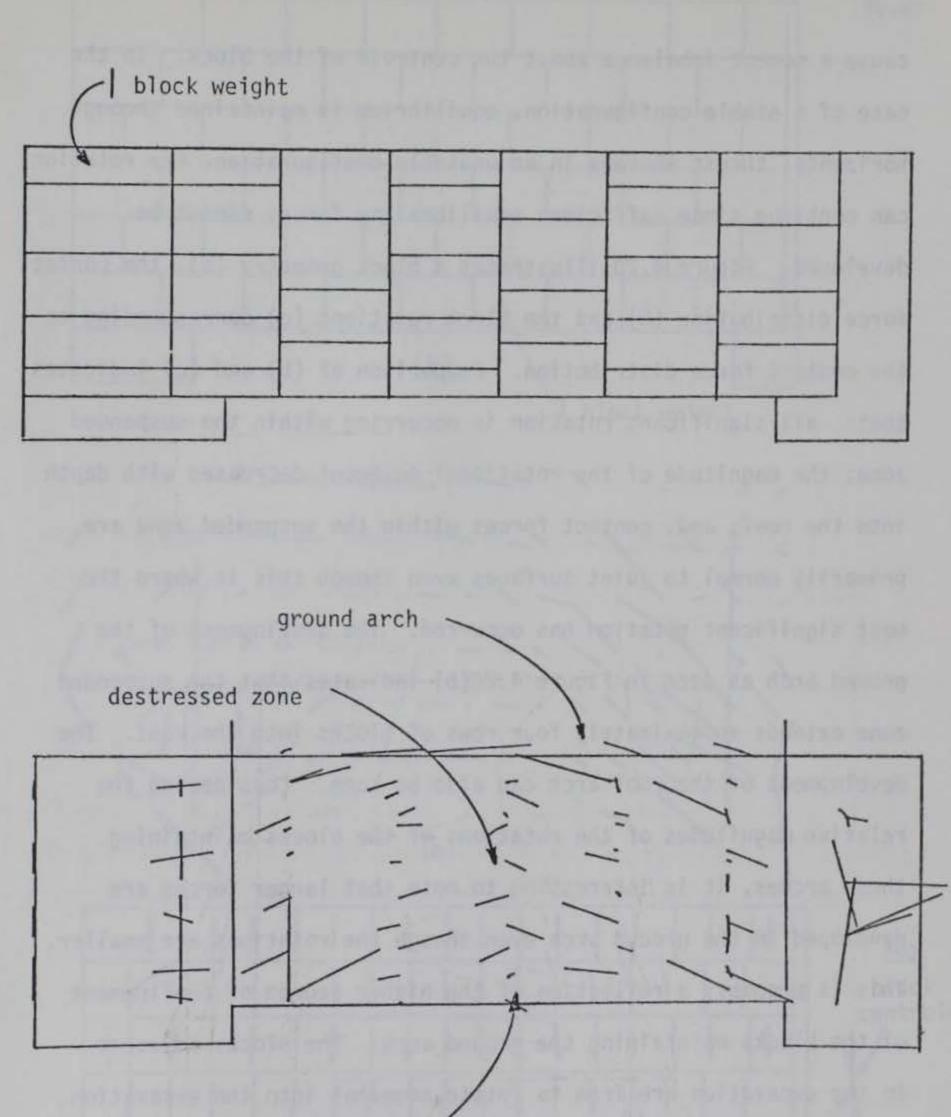


Figure 4.18 Formation of ground and roof arches in a continuously jointed model.



IV-45

roof arch

Figure 4.19 Formation of the ground and roof arches in a horizontally discontinuous jointed model.

cause a moment imbalance about the centroid of the block. In the case of a stable configuration, equilibrium is maintained through horizontal thrust whereas in an unstable configuration, the rotation can continue since sufficient equilibrating forces cannot be developed. Figure 4.20 illustrates a block geometry (a), the contact force distribution (b) and the block rotations (c) corresponding to the contact force distribution. Comparison of (b) and (c) indicates that: all significant rotation is occurring within the suspended zone; the magnitude of the rotational movement decreases with depth into the roof; and, contact forces within the suspended zone are primarily normal to joint surfaces even though this is where the most significant rotation has occurred. The development of the ground arch as seen in Figure 4.20(b) indicates that the suspended zone extends approximately four rows of blocks into the roof. The development of the roof arch can also be seen. Considering the relative magnitudes of the rotations of the blocks maintaining these arches, it is interesting to note that larger forces are developed in the ground arch even though the rotations are smaller. This is probably a reflection of the higher degree of confinement of the blocks maintaining the ground arch. The blocks adjacent to the excavation are free to rotate somewhat into the excavation.

The next row of blocks upward thus has the freedom to rotate toward the excavation although not as much as the lower row. Successively less rotation is permitted until at the limit of the suspended zone, minimal rotation is occurring.

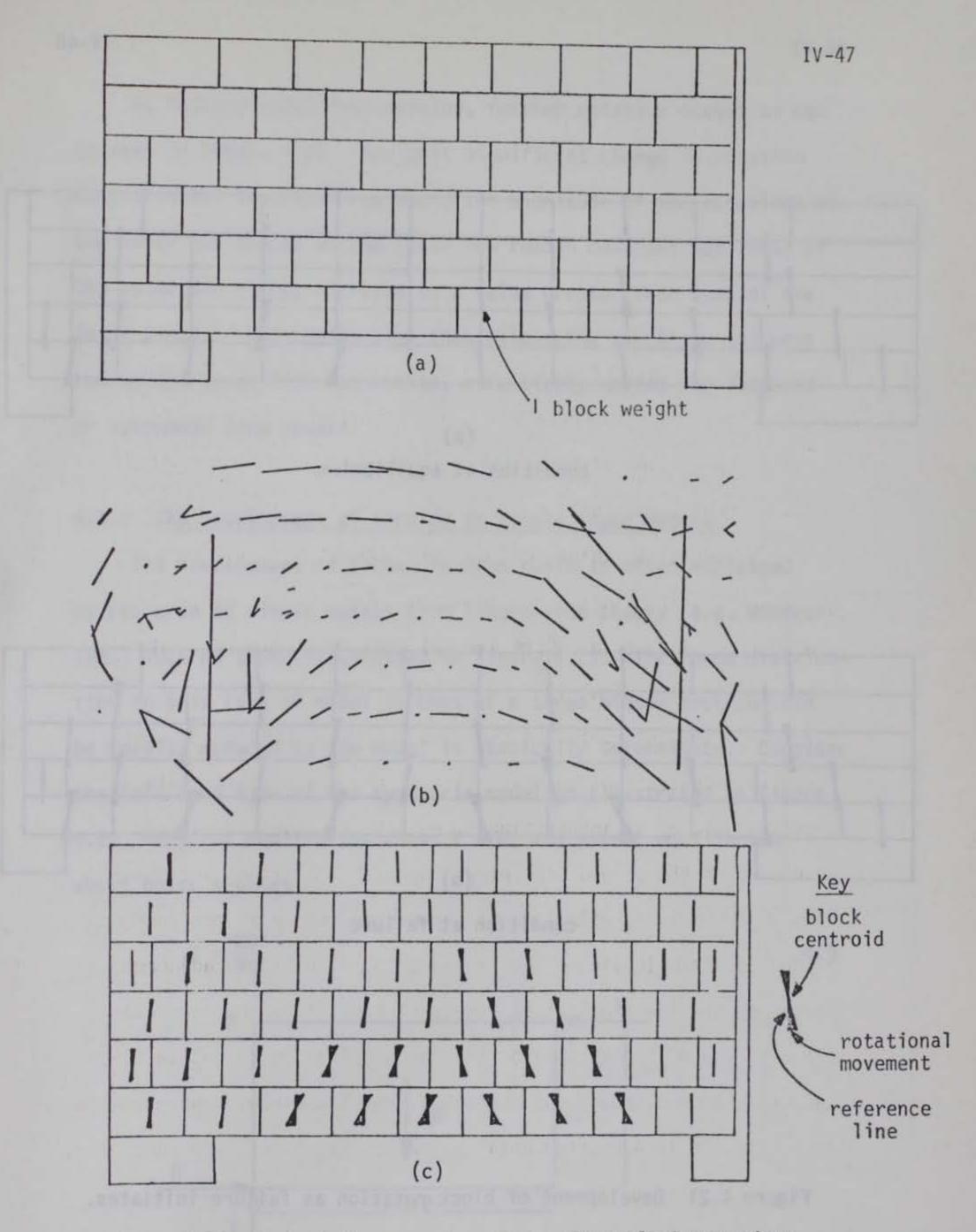
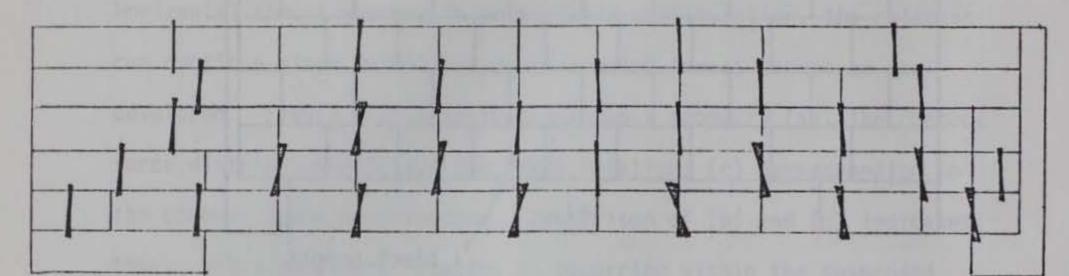


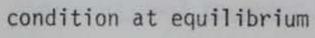
Figure 4.20 Contact forces and corresponding block rotations.

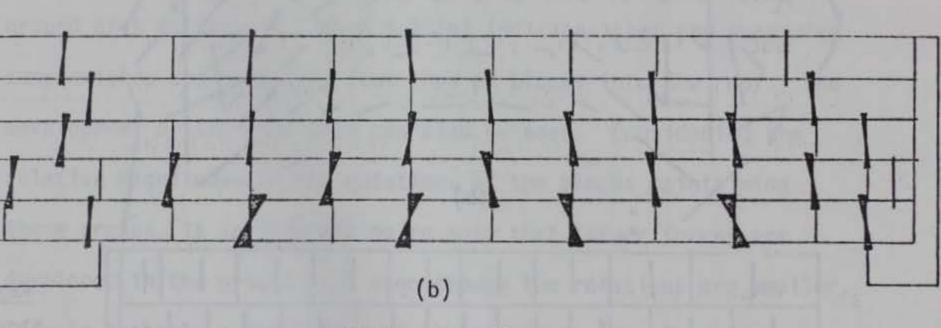
TALKS A ALL THE LINEAR SPACE PROPERTY





(a)





condition at failure

Figure 4.21 Development of block rotation as failure initiates.

iligare d. St. Concert formers and steresponding block cotations.

As failure conditions develop, further rotation occurs as can be seen in Figure 4.21. The most significant change in rotation occurs in the lowermost row where the magnitude of the rotations of the inner two blocks of the lower row remain constant but those of the outer two blocks increase to a value greater than that of the inner blocks. This deflection then allows the blocks in the next row upward to deflect and rotate, effectively moving the loosened or suspended zone upward.

4.5.3 The development of arching in single layer models

The development of arches in mine roofs is often explained by recourse to simple models from linear arch theory (e.g. Woodruff, 1966) such as those illustrated in Figure 4.23. The force distribution in this type of model is that of a three hinged arch and can be readily deduced as the model is statically determinate. Consider the left hand side of the symmetric model as illustrated in Figure 4.22, vertical equilibrium shows V = W, and moment equilibrium about point a shows:

$$H = \frac{WC}{4t}$$

a

H

4.9

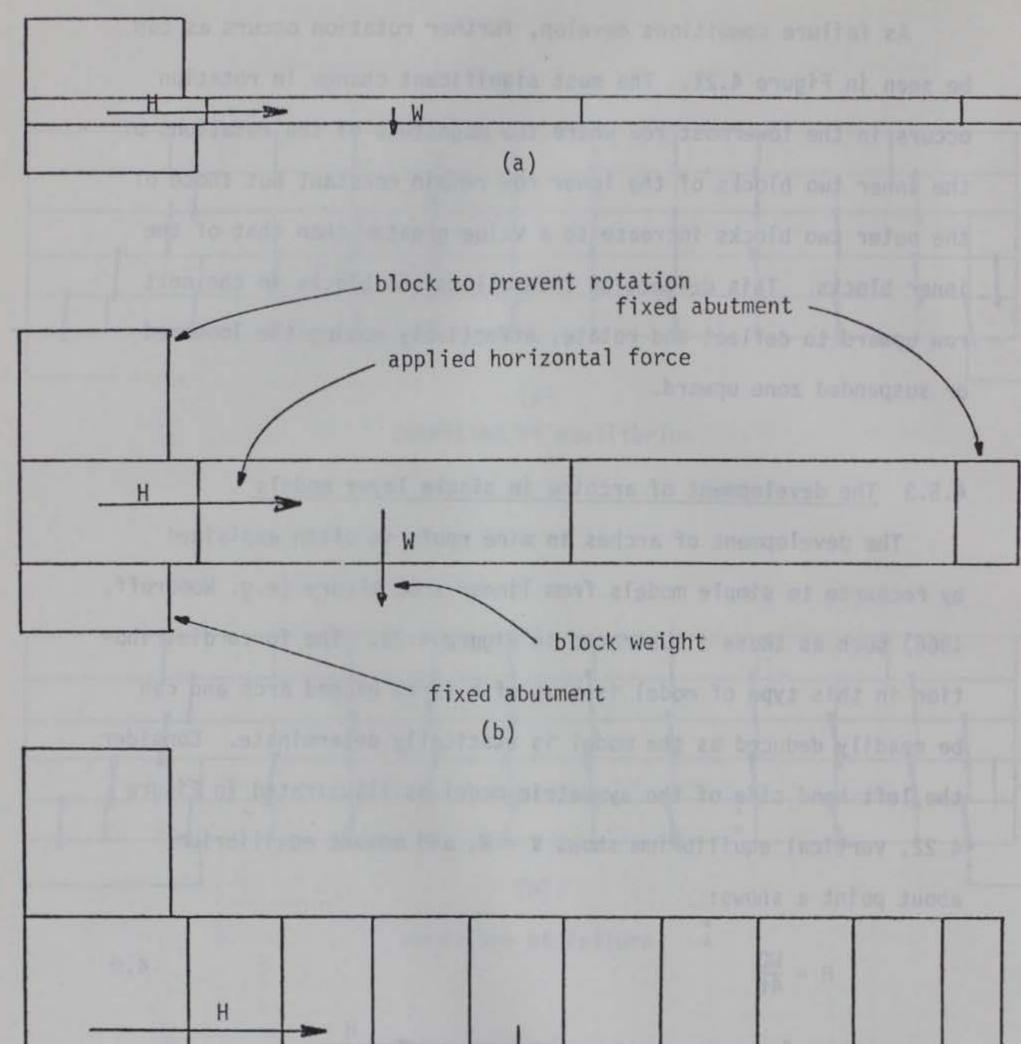
The Linear Arch Model Figure 4.22

0/2

W

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RAST



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(c)

Figure 4.23 Typical block models for linear arching study.

This force distribution represents a limiting condition; as vertical deflection of the beam causes the contact at the lower face to be broken, the value of the lever arm t decreases and thus an increasing value of H is required for stability.

Analyses by the Distinct Element method of several linear arch models is summarized in Table 4.1 and indicates that Equation 4.9 may be used to predict the horizontal thrust required for stability in certain instances. These data show that equation 4.9 is correct for low aspect ratios of the blocks but loses validity as block thicknesses increase and friction coefficients of the joints decrease. For larger block thicknesses and lower friction coefficients, the horizontal thrust required for stability is found accurately by equation 4.3 which is repeated here for convenience:

$$H = W/2 \cot \phi$$
 4.3

Analysis of the force distribution at failure provides insight into this discrepancy. Figure 4.24 illustrates the force distribution at failure in models C, A and D. Figure 4.23(a) illustrates conditions at failure for model C with μ = 0.5. Full frictional resistance is mobilized on the abutment joints and compression is transmitted across the lower contact of the mid span joint. Although arching

is developing, failure is by sliding along the abutment joints. Figure 4.24(b) illustrates the force distribution for model A with $\mu = 1.0$. Arching is fully developed as evidenced by the absence of force transmittal at the lower mid span joint contact. An important distinction in this case is the fact that frictional resistance is 1 CARLOR

Table 4.1 Summary of Linear Arch Models

at the proof which and the process and in which the

	Friction Coefficient	Predicted Load		Observed	Observed Failure Mode	
Model	μ	Arching ⁴	Sliding	Side Load at Failure		
	.25	500	280	500 2	Arching	
A 1	.5	500	140	500	Arching	
	1.0	500	70	500	Arching	
	.25	500	550	550 3	Sliding	
В	.5	500	280	500	Arching	
	1.0	500	140	500	Arching	
brts	.25	500	1120	1110	Sliding	
С	.5	500	560	550	Sliding	
- C.K	1.0	500	280	490	Arching	
	.25	500	2580	2550	Sliding	
D	.5	500	650	650	Sliding	
	Lindhetzib er	and the second	south is	a man in	any methode	

UNDER LOOS CONFLICTED CANED & TOTAL

Notes: 1 Geometry of models

Model A t = 25, 0 = 700, 2 block linear arch model Model B t = 50, 0 = 700, 2 block linear arch model Model C t =100, 0 = 700, 2 block linear arch model Model D t =225, 0 = 700, 8 block, voussior beam
2 Difference in calculated side load for arching models is typically less than 2%.

3 Difference in calculated load for sliding models is typically less than 1%.

4 Equation 4.1 may be rewritten by recognizing that W is a function of t and O (W = t $\times \frac{0}{2} \times d$); substitution leads to (density, d = 1) H = $\frac{0^2}{8}$ and thrust is thus independent of block thickness.

12 24

(a)

(b)

(c) Figure 4.24 Force distributions in linear arch model (force scale

Figure 4.24 Force distributions in linear arch model (force scale from Figure 4.23).

not fully developed along the abutment joints. The vertical component of the abutment reaction is equal to the weight of the roof block while the horizontal component is equal to the horizontal thrust required to maintain stability against arching (equation 4.9).

This fact permits the calculation of the critical friction coefficient that delineates arching failure from frictional sliding in the linear arch model. Consider an opening of span O, with the roof blocks having thickness t, and weight W per block. From linear arch theory, the thrust developed during arching is:

$H = \frac{WO}{4t}$ 4.9

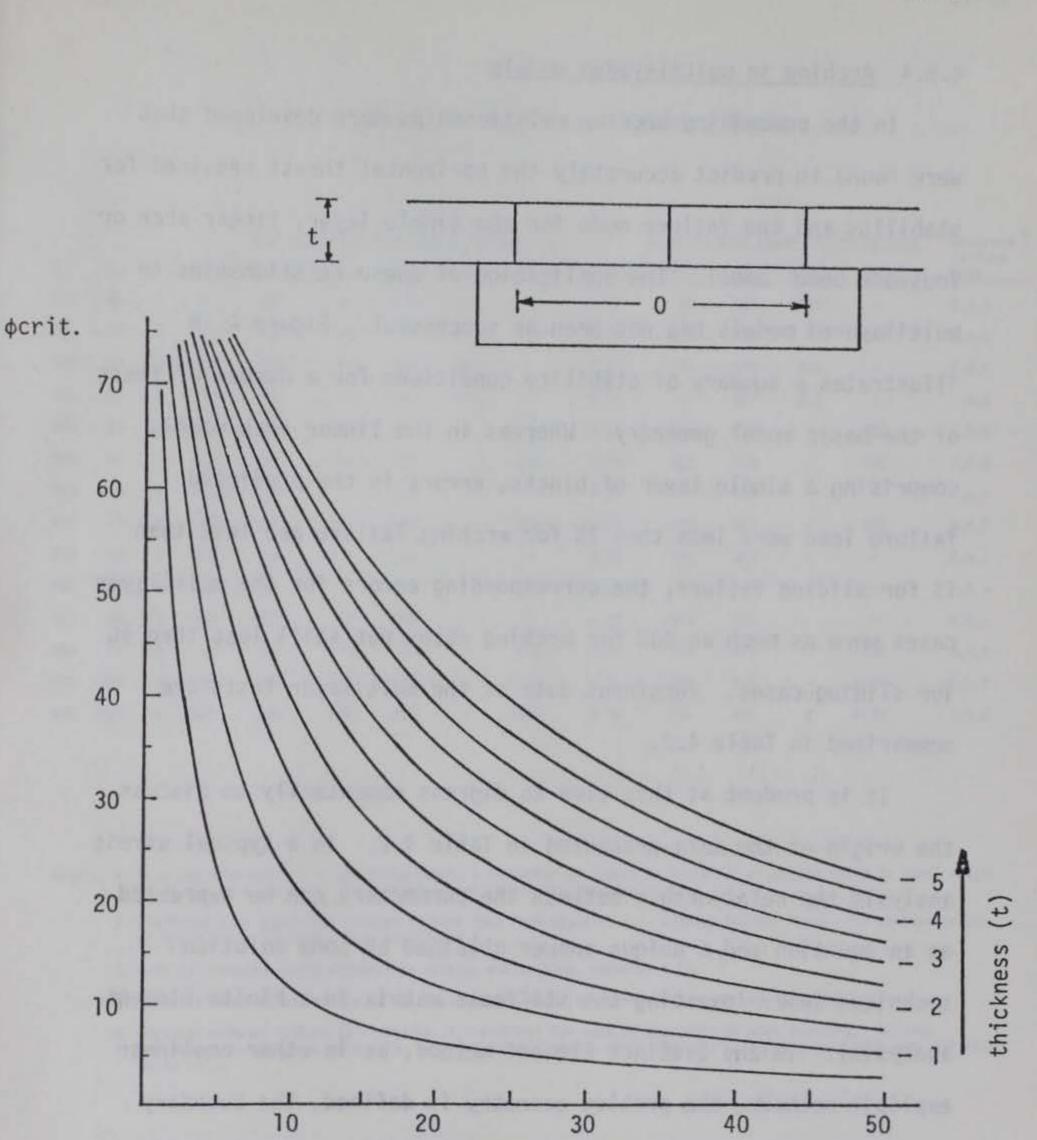
The critical friction angle (ϕ crit) is the inverse tangent of the ratio of the block weight and the thrust force:

$$\phi \operatorname{crit} = \tan^{-1}\left(\frac{4t}{0}\right)$$
4.10

If the friction angle of the joints is greater than this critical value, sliding cannot occur and failure, if it occurs, will be by true arching. On the other hand, if the friction coefficient on the joints is less than this critical value, sufficient frictional resistance cannot be developed and failure occurs by sliding.

Equation 4.10 is plotted in Figure 4.25; this figure may be

used to determine if, for a given span and block thickness, failure will be by true arching or by slippage with only partial development of arching conditions. The equation has been found to be correct for all linear arch models analyzed.



IV-55

Span (0)

Figure 4.25 Critical friction angle as a function of excavation span and block thickness (span and thickness must be in consistent units).

4.5.4 Arching in multilayered models

In the preceeding section relationships were developed that were found to predict accurately the horizontal thrust required for stability and the failure mode for the single layer, linear arch or Voussoir beam model. The application of these relationships to multilayered models has not been as successful. Figure 4.26 illustrates a summary of stability conditions for a number of tests of the basic model geometry. Whereas in the linear arch model, comprising a single layer of blocks, errors in the predicted failure load were less than 2% for arching failure and less than 1% for sliding failure, the corresponding errors for the multilayer cases were as much as 40% for arching cases but still less than 1% for sliding cases. Pertinent data of the multilayer tests are summarized in Table 4.2.

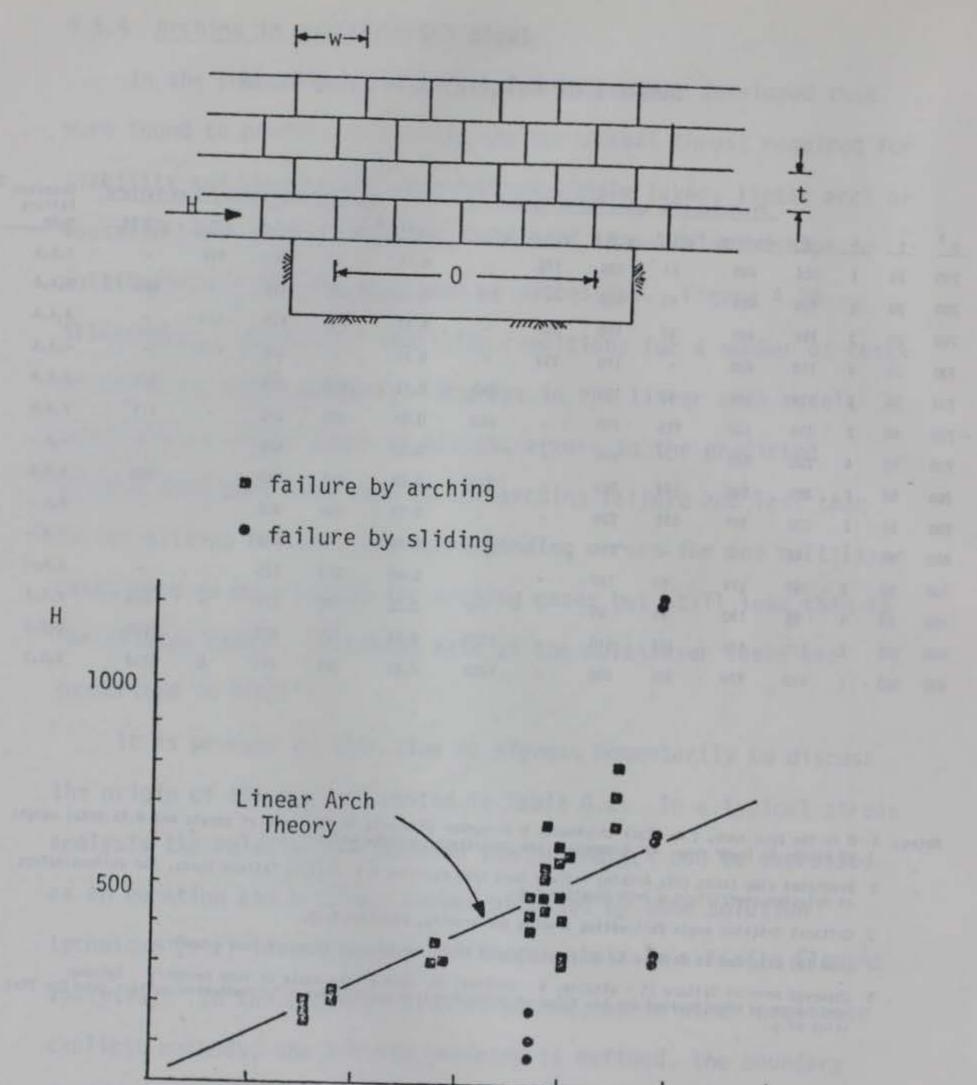
It is prudent at this time to digress momentarily to discuss the origin of the data presented in Table 4.2. In a typical stress analysis the relationship between the parameters can be expressed as an equation and a unique answer obtained by some solution technique (viz. inverting the stiffness matrix in a Finite Element analysis). In the Distinct Element method, as in other nonlinear explicit methods, the problem geometry is defined, the boundary conditions are specified and subsequent motion of the blocks is observed; equilibrium occurs as the force distribution converges to a situation where the relative accelerations of the blocks approaches zero. In terms of the problem at hand this means that a set of

Table 4.2

Summary of Multilayer Arching Tests

				Predicted Side Loads (H) at Failure 2				Observed Side Loads (H) at Failure ⁴				Observed 5		
01	_ <u>t</u> _	<u>_b</u> _	_ <u>W</u>	Arching	<u>p=1.0</u>	<u>µ=0.5</u>	<u>µ=0.3</u>	µ=0.25	¢ crit 3	<u>µ=1.0</u>	<u>µ=0.5</u>	<u>µ=0.3</u>	µ=0.25	Failure Mode
700	20	1	106	460	53	106	176	-	0.11	55	105	175	-	\$,\$,\$
700	20	2	106	460	53	105	-	212	0.11	385	425		465	A,A,A
700	20	3	110	480	55	110	185	-	0.11	440	470	515	-	A,A,A
700	20	4	110	480	-	110	193		0.11	-	540	650	-	-,A,A
750	20	б	120	560	60	120	-	240	0.11	650	725	-	800	A,A,A
700	40	2	230	500	115	230	-	460	0.23	300	315		415	A,A,A
700	50	4	290	420	-	290	-		0.29	-	575	-	-	-,A,-
700	50	2	285	500	143	285	-	570	0.29	475	560	-	600	A,A,A
600	50	2	230	345	115	230	-	-	0.33	300	350	-	-	A,A,-
600	40	4	196	360	-	196	-		0.25	-	300	-	-	-,A,-
500	50	2	180	225	90	180		-	0.40	200	225	-		A,A,-
450	25	4	85	190	43	85	-	170	0.22	150	175		200	A,A,A
800	100	2	610	570	305	610		1220	0.50	325	625	-	1225	s,s,s
800	100	1	610	570	305	600	•	1220	0.50	305	615	0	1210	\$,\$,\$

- Notes: 1 O is the true span, t is block thickness, b is number of blocks in lower row of strata and W is total weight of blocks in lower row. All dimensions are consistent computer units.
 - 2 Predicted side loads (H): Arching failure load from equation 4.9, Sliding failure loads, for various values of friction coefficient u from equation 4.6.
 - 3 Critical friction angle delineating sliding and arching, equation 4.10.
 - 4 Load (H) observed at failure in Distinct Element model for several tests of same geometry.
 - 5 Observed mode of failure (S sliding, A arching) for each of the tests of same geometry. Columns correspond to high, medium and low value of joint friction coefficient. "-" indicates no test data for that value of µ.



1000 2000 3000 4000 <u>Ow</u> t

Figure 4.26 Summary of multilayer arching tests (all dimensions in computer units).

boundary conditions is applied and the program allowed to run until it is determined that the geometry is stable. The boundary conditions are then incrementally modified and again the program is allowed to run. This iteration is then continued until failure occurs. Thus, each data point on Figure 4.26 represents a limiting condition deduced by a minimum of four or five computer runs.

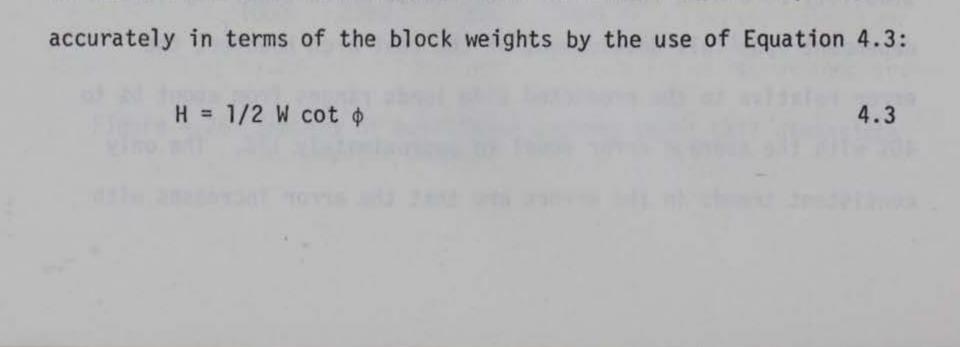
The problem of determining equilibrium conditions is discussed further in Appendix B.

Tabulated in Table 4.2 are predicted side loads for stability obtained by Equation 4.9 for arching conditions and by Equation 4.6 for sliding conditions. The observed loads at failure are also tabulated and comparison indicates a general divergence from the predicted values. Nine of the tests developed sliding failure modes and are indicated by a circular symbol in the plot of Figure 4.26; the remainder of the tests developed full arching failure modes and the data points are seen to follow the general trend of the linear arch model as represented on Figure 4.26 by the square symbols.

In those tests where failure was by frictional slippage, the side loads were typically within 2% of the value predicted by Equation 4.6; the indication being that in those cases where full arching does not develop, Equation 4.6 may be used to assess the stability of a mine roof. For those tests where stability is dependent upon full development of the roof arch however, the error relative to the predicted side loads ranges from about 5% to 40% with the average error equal to approximately 17%. The only consistent trends in the errors are that the error increases with the number of blocks in the lower row and that for a fixed geometry the error either increases or moves from negative to positive as the friction angle increases.

Analysis of the linear arch, single row models led to the calculation of a critical friction angle (Equation 4.10) that was found to predict accurately the dividing line between failure by arching and failure by sliding along the abutment joints. The tangent of the critical friction angle for each of the multilayered block tests is also tabulated in Table 4.2; several instances can be found in the table which illustrate discrepancies between actual and predicted failure modes with arching failure modes developing in several instances where the critical friction angle concept predicted a sliding failure mode.

Examination of the data indicates that failure by full development of the roof arch is more likely to occur than failure by sliding along the abutment joints. Exceptions to this observation were found only in those instances where the development of the arch was somehow constrained. Specific conditions that lead to failure by slippage were the expected case where the main roof was monolithic and arching could not develop, and cases where the block thickness was relatively large and the main roof comprised only two blocks. In these instances the horizontal load at failure could be predicted



The most noticeable departure from the observed behavior of the single layer linear arch models was concerned with contact force distribution along the lower row of blocks. In the single layer models, failure always initiated as the central contact along the lower face opened; as noted earlier, this was the expected behavior since the deflection of the blocks reduced the moment arm of the horizontal stabilizing force resulting in increasingly unstable conditions. This phenomonon is, however, not indicative of the behavior of the multilayer models.

The conditions preceeding failure in the multilayer models are characterized by two common features. First, loss of force transmittal across the lower contact of the midspan joint is not indicative of failure. Frequently, significant horizontal force reduction after the joint opens is required before failure occurs. The second general behavior pattern that was recognized concerns the distribution of contact forces in the immediate roof. Figure 4.27 presents a typical multilayer model and a section of its contact force distribution. The blocks are in equilibrium but a reduction in the horizontal thrust of approximately 10% would lead to failure; this is a typical force distribution of a multilayer model at stress conditions slightly greater than those at which failure occurs.

Three characteristics of the force distribution in multilayer models have been noted in all models tested and are indicated in Figure 4.27 by the letters A, B, and C. The characteristics are: A) absence of force transmittal across the lower contact of the mid span joint

- B) minimal vertical transmittal within the suspended zone, especially to the lower row of blocks
- C) the development of an additional contact force where the blocks adjacent to the abutment rotate into the next upward level of blocks

The second characteristic is to be expected in light of the model; the corbelling effect of the blocks outside of the suspended zone acts to lessen the span over which the next row of blocks must be supported. In this particular case, the span is decreased by 25%, the weight to be supported is decreased by 25% and the required horizontal force to just maintain equilibrium is 45% of that which is actually being applied. This simple calculation neglects the vertical force transmittal which is occuring to the second row of blocks, but the fact that the thrust applied to the second row of blocks is almost twice that required for stability indicates why the deflection of the second row is small compared to that of the lower row and thus why no vertical force transmittal occurs to the lower row.

The other two observations, A and C, are closely related and provide a reasonable explanation as to why the behavior of the multilayer models depart from the linear arch model. Figure 4.28 is a schematic representation of the two blocks on the left hand

side of the lower row of blocks in Figure 4.27(a) based on the contact force distribution of Figure 4.27(b). The linear arch model is based upon the contact force distribution illustrated in Figure 4.22; comparison of these two figures indicates that the model used

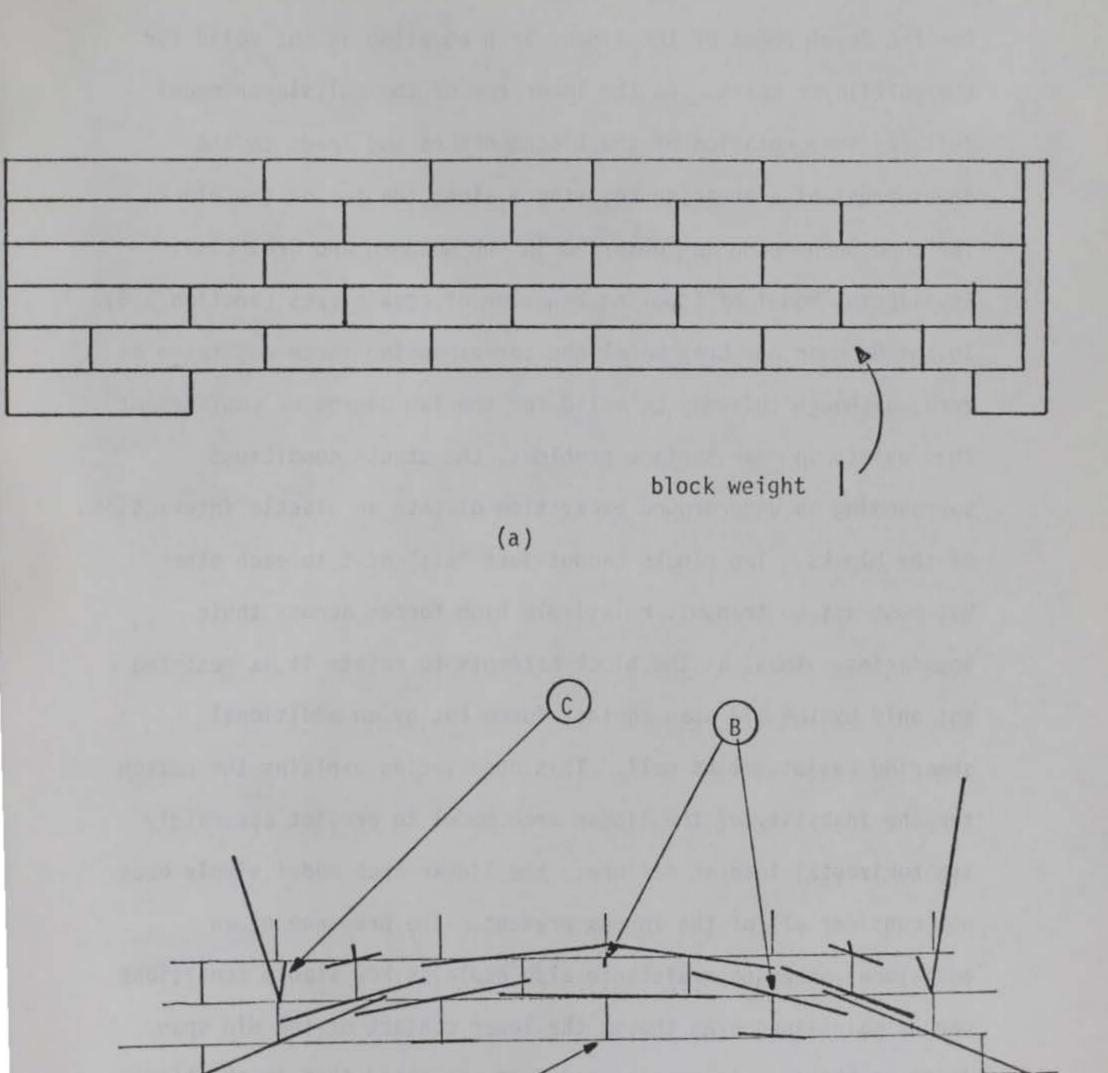


Figure 4.27 Contact force distribution in lower rows of multilayer model.

(b)

for the development of the linear arch equation is not valid for the multilayer cases. As the lower row of the multilayer model deflects some rotation of the blocks occurs and leads to the development of a shearing resistance along the top of the block. The same phenomenon was observed in the Goodman and Bray Limit Equilibrium Model of toppling behavior of rock slopes (section 3.6). In the Goodman and Bray model the corresponding force was taken as zero; although this may be valid for the low degree of confinement that exists in near surface problems, the stress conditions surrounding an underground excavation dictate an elastic interaction of the blocks. Two blocks cannot just "sit" next to each other but must act to transmit relatively high forces across their boundaries. Thus, as the block attempts to rotate it is resisted not only by the mid span contact force but by an additional shearing resistance as well. This observation explains the reason for the inability of the linear arch model to predict accurately the horizontal load at failure: the linear arch model simply does not consider all of the forces present. The presence of an additional shearing resistance also explains how stable conditions can be maintained even though the lower contact of the mid span joint is broken. In section 4.3.5 it was noted that in the linear

arch model, once this contact opened, the governing equation dictated that failure must occur. The presence of the additional force acting on the block tends to maintain equilibrium in a manner not accounted for by the linear arch model. Unlike the linear arch model, the force distribution presented

in Figure 4.28 is statically indeterminate. To develop an equation relating span, block thickness, joint spacing, block weights and friction coefficient would require that two assumptions be made concerning the forces. The logical assumptions would be to assume the development of full frictional resistance of the two contacts experiencing shear. However, in the majority of tests run, full frictional resistance was not seen to develop at either contact. Rather, the Distinct Element method can be used to study each model on an individual basis and develop relationships not subject to arbitrary assumptions regarding the force distributions.

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Figure 4.28 Force distribution observed during arching in multilayer models.

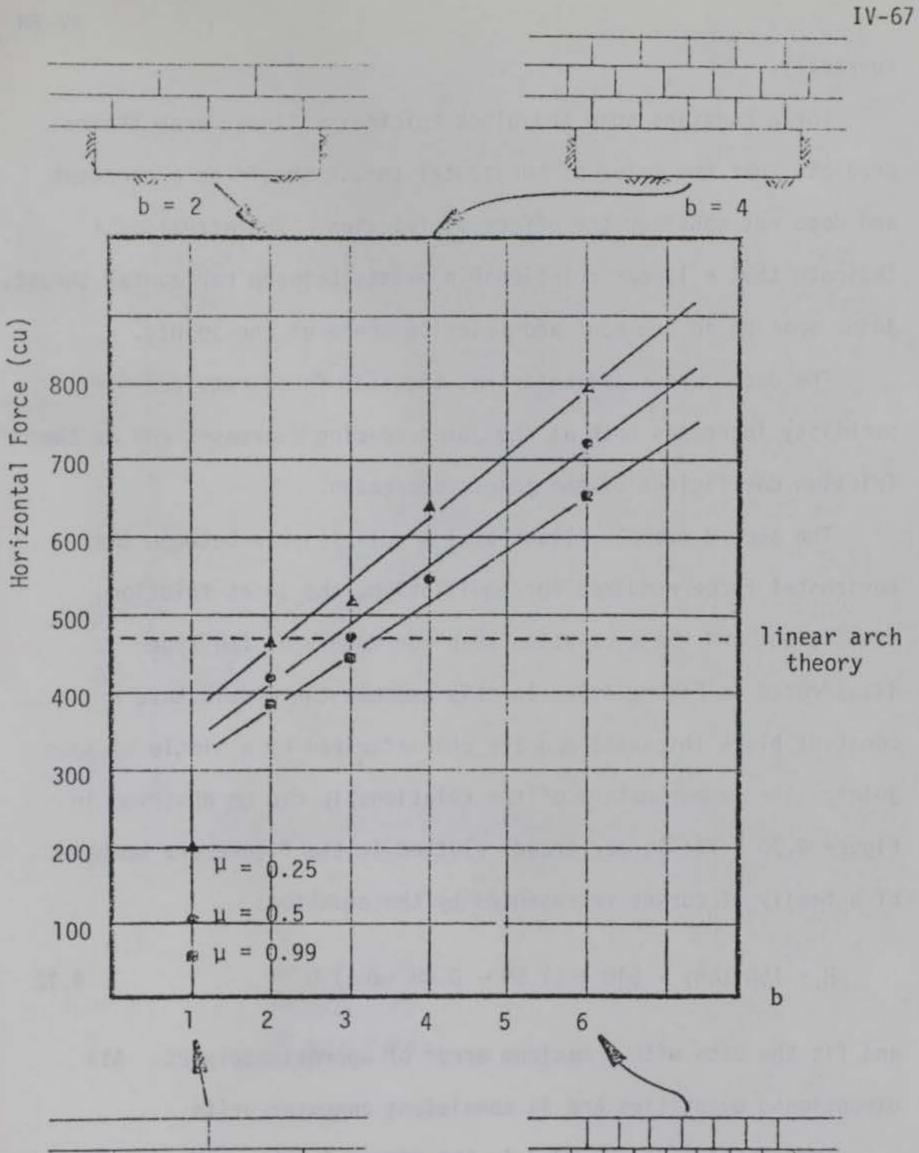
4.6 Use of Results in Design

The results from the previous Distinct Element runs can be expressed in a way that may be useful for design purposes. The two examples presented below utilize the data of Table 4.2 to derive empirical relationship between parameters. These relationships are characterized by errors in the order of 4% rather than the 40% error experienced when using linear arch theory to predict the horizontal thrust.

The first example derives a relationship between the horizontal force required for stability, the number of blocks in the bottom row, (a factor which is analogous to joint spacing) and the friction angle of the joints, in models similar to those shown in Figure 4.3. The excavation width and the block thickness are constant in this analysis. The data points, which represent the failure conditions for 11 test models, and the associated linear trends are plotted in Figure 4.29. The linear trends in the figure are members of a family of curves represented by the equation

H = 314.3 - 59.5 tan\$\phi + (87.3 - 19.3 tan\$\phi) b 4.11
with all dimensions expressed in consistent computer units. Also
included in the figure is a horizontal dashed line which represents

the value of horizontal force necessary to maintain roof stability as calculated by linear arch theory. The data points corresponding to a monolithic lower roof (b = 1) are included on the plot and are seen to deviate from the trend of Equation 4.11; the frictional resistance relationship (Equation 4.6) predicts these values



Nº MA b = 1TIN 11115 b = 6Figure 4.29 Linear relationship between horizontal force, number of blocks in the lower row and joint friction angle (constant span and block thickness).

correctly.

For a constant span and block thickness, linear arch theory predicts that the value of horizontal thrust should be a constant and does not consider the effect of friction. The actual data indicate that a linear relationship exists between horizontal thrust, joint spacing in the roof and friction angle of the joints.

The data values indicate that the side force required for stability increases both as the joint spacing decreases and as the friction coefficient of the joints decreases.

The second example illustrates a relationship between the horizontal force required for equilibrium, the joint friction coefficient and the excavation span for models of the type illustrated in Figure 4.3. In this example the models have a constant block thickness and are characterized by a single midspan joint. The linear nature of the relationship can be observed in Figure 4.30. The linear trends plotted in the figure are members of a family of curves represented by the equation:

 $H = 190 \tan \phi - 540 + (1.59 - 0.48 \tan \phi) 0$ 4.12

and fit the data with a maximum error of approximately 2%. All dimensioned quantities are in consistent computer units.

The dashed line included in the figure is the value of side load predicted by linear arch theory. The required horizontal force for stability is seen to increase with span as predicted by linear arch theory but the linear arch theory does not take account of the fact that an increase in the joint friction angle reduces the horizontal

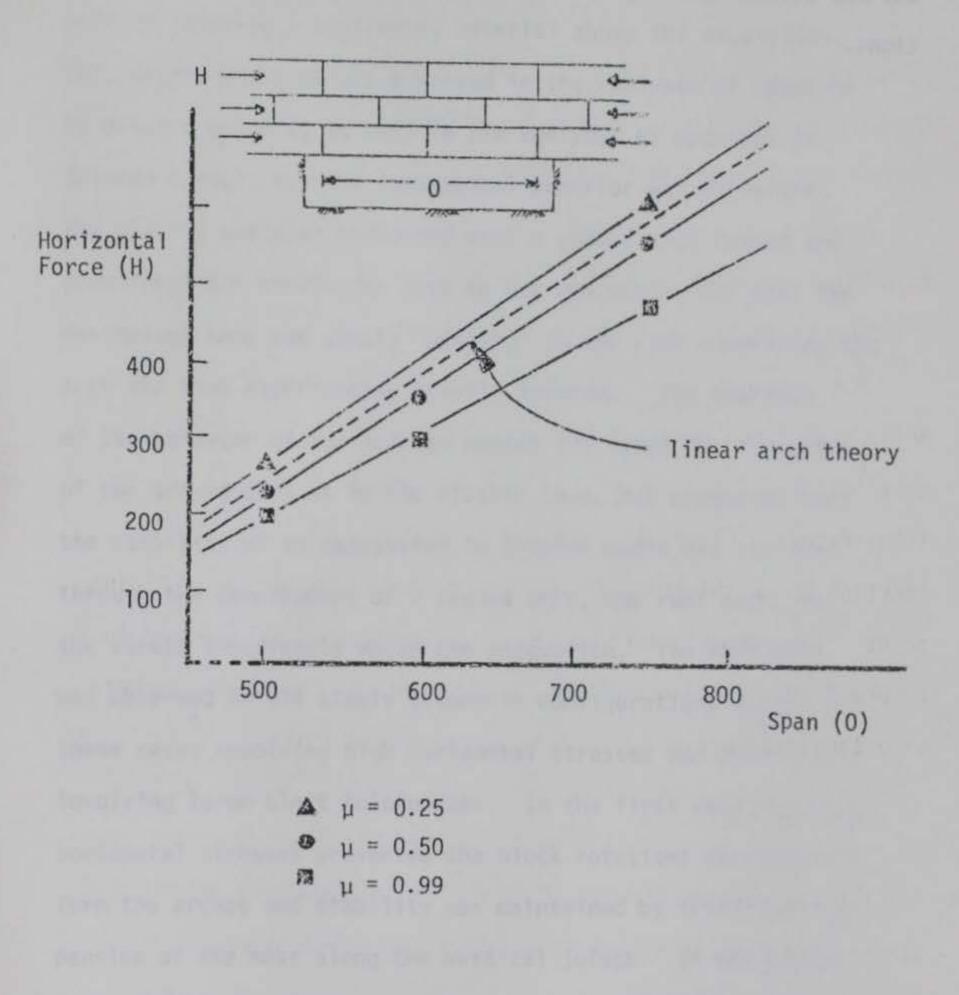


Figure 4.30 Linear relationship between span, horizontal force and joint friction angle (constant block thickness and one midspan joint; all dimensions in computer units).

where finding these time transfer to be been a working the set of the set

load required for stability. This reduction is due primarily to the additional shearing resistance provided by the layer interactions.

4.7 Summary

The stability of excavations in jointed rock was seen to be governed by mechanisms of stress transfer which resulted in a zone of relatively destressed material above the excavation. This destressed zone was observed in the analyses of openings in elastic material as well in the analyses of openings in jointed masses, but the fundamental behavior was different. The elastic analyses indicated that a ground arch formed and transfered the overburden load to the abutments, but that the destressed zone was simply "hanging" on the rock comprising the arch and thus experiencing tensile stresses. The analyses of the behavior of the jointed masses indicated the formation of the ground arch as in the elastic case, but suggested that the stability of an excavation in jointed media was attained through the development of a second arch, the roof arch, in the strata immediately above the excavation. The roof arch was observed in all stable geometric configurations except for those cases involving high horizontal stresses and those cases involving large block thicknesses. In the first case the high horizontal stresses prevented the block rotations necessary to form the arches and stability was maintained by frictional suspension of the mass along the vertical joints. In the second

case, the block thickness, relative to the excavation span, reached a point at which the arch development was constrained and failure of the mass was by sliding along the joints. It was found that the transition between arching and sliding behavior could be predicted accurately. The Distinct Element obtained solutions for single layer, self loaded, jointed beams were compared to a linear arch theory neglecting the compressive strength of the rock and the lateral stiffness of the abutments; agreement of the data with theory was quite good. When the single layer, linear arch theory was compared to multiple layered models, however, agreement of the data and theory was poor. The discrepancy was seen to be due to layer interactions, not accounted for in the single layer model, acting in a manner that increased the horizontal thrust on the abutments.

A Limit Equilibrium solution for the observed contact force distribution was calculated, but discarded since the contact vectors were seldom observed to be at fully developed frictional resistance. Instead, the data was examined in order that the significant parameters and the relationships between them could be isolated. Two main conclusions could be drawn from the data. First, there is a linear relationship between the span and the horizontal thrust required for stability of the mass. However, in contrast to linear arch theory, the models examined by the Distinct Element method indicated that this relationship involved the joint friction coefficient. This was observed to be due to interactions between the lower two layers and not a resultant of slipping along the vertical joints at the abutments.

The second identified relationship indicated that the horizontal thrust was a function of the joint spacing, expressed as the number of blocks in the lower row of strata, and the joint friction coefficient. The significance of this observation lies in the fact that linear arch theory does not account for an effect due to joint spacing. The data indicate that as the number of blocks in the lower row of strata increases from two to six, the horizontal stress required for stability almost doubles; linear arch theory, on the other hand, predicts that this horizontal stress should be a constant value.

To keep a proper perspective, it must be noted that the analyses described in this chapter were performed with a restricted behavior model possessing infinite strength and regular jointing. More sophisticated linear arch theories account for load transfer between layers and the compressive strength of the material. The real situation in bedded roofs involves crushing of the rock which can change the length of the moment arm used to calculate the horizontal thrust in the linear arch theory. It must be concluded that it may be invalid to criticize linear arch theory or the basis of the analyses just described. The analyses do indicate, however, that mechanisms act in jointed rock that perhaps should be implemented in a comprehensive linear arch theory.

CHAPTER V

AN ANALYSIS OF SUPPORT REQUIREMENTS OF EXCAVATIONS IN JOINTED ROCK MASSES

5.1 Introduction

In a historical review of tunnel construction, Szechy (1970) states that the oldest known tunnel other than those associated with mines is, according to present knowledge, over 4000 years old. This tunnel was constructed in Babalonia during the reign of Queen Semiramis to underpass the River Euphrates. The length of this tunnel was over 1 km and it had a cross-section of 3.6 m by 4.5 m. Although built by cut and cover methods, elements of the structure demonstrated (viz. a vaulted arch for the roof) that the Babylonians possessed considerable skill in tunnel construction, most likely gained from experience in previous tunneling ventures. To fully emphasize the significance of this undertaking, Szechy notes that it wasn't until 1843 that the next subaqueous tunnel, that crossing the River Thames in London, was opened, almost 4000 years later.

Significant increases in the magnitude of the scale of projects typically undertaken in underground excavation have not been accompanied by, or for that matter, preceeded by analytical techniques capable of explaining the complex behavior of the structural system

comprising the rock mass and the support system. The design of

tunnel or excavation support systems are routinely guided by

empirical and observational rock load prediction schemes. It is universally acknowledged that the use of these schemes results in

an overdesign, but the majority of research undertaken today seems not to be directed toward understanding the mechanisms responsible for the behavior of an excavation but toward somehow strengthening the position of the empirical methods through the acquisition of additional data. This approach has helped to identify the parameters to which support design is most sensitive, but the fact that excavation support design is highly site dependent does not obviate the need for rational methods for the prediction of support pressures. This chapter presents the results of analyses of jointed rock masses which utilize the Distinct Element method to characterize the interaction of a jointed rock mass with a support system. The vehicle chosen to quantitatively express this interaction is a ground reaction curve. A ground reaction curve is simply a plot of the support force necessary to maintain the stability of a rock mass as a function of displacement of the rock mass. The utility of the ground reaction curve in support design is that it typically yields

information about the optimum time of support emplacement as well as the magnitude of the force the supports must resist.

Previously, ground reaction curves have only been calculated by continuum based methods; the rock was assumed to be broken but the representation of the behavior was by a plastic or elasticplastic constitutive relationship.

V-2

The Distinct Element formulation provides the research tool

necessary to investigate load-deflection relationships in a medium

where the deformation is controlled solely by the jointing. The

ground reaction curves presented in this chapter indicate a

relationship between required support force and the geometric parameters defined by the excavation dimensions and the joint spacings. This data was also compared to predictions made by several of the empirical methods in an attempt to determine if any correlation could be found.

5.2 The Estimation of Rock Loads for Support Design

5.2.1 The concept of a ground reaction curve

As an introduction to the discussion of the various methods commonly in use to design reinforcement schemes in tunnels it is prudent to discuss a theoretical concept which provides a means to quantitatively describe the behavior of the rock mass as it is disturbed by an excavation. This concept is concerned with the interaction of the material surrounding the excavation and the support system emplaced to ensure stability. The behavior of the material is described by a ground reaction curve relating the force required to stabilize the mass to the deformation of the edge of the excavation. As an illustration of the concept, an example (Deere et al., 1969) describing a ground reaction curve for a soil mass is presented.

The basis for establishing the stress for which a tunnel lining should be designed is illustrated in Figure 5.1 where the average radial stress on a circular tunnel lining is plotted as a function of the average inward radial deformation of the tunnel wall. The point A illustrated in the figure represents the average radial stress before excavation occurs.

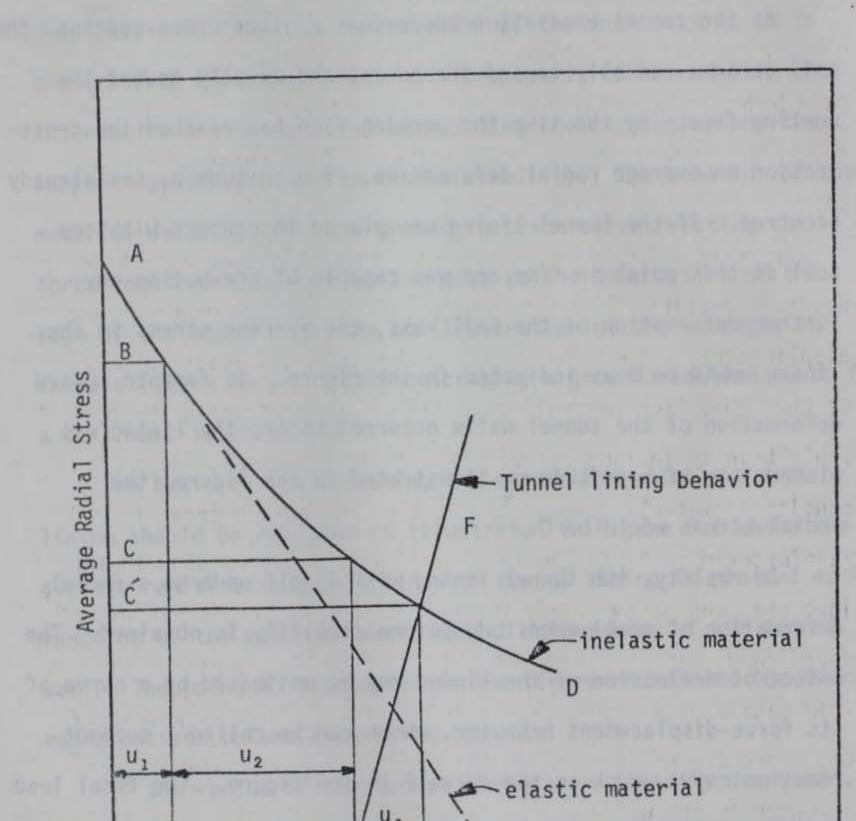
If the radius of the tunnel lining were steadily decreased, the

load on the tunnel lining would decrease in accordance with a relationship describing the stress-strain-time characteristics of the soil. If the soil were elastic the relationship would be linear as shown in the figure by the dashed line AE; for the more likely case that the material is inelastic, the relationship could resemble the curve AD. This relationship is termed the ground reaction curve. The form of the ground reaction curve cannot be calculated exactly but may be approximated in several instances of practical importance on the basis of field observations coupled with theoretical investigations.

As the tunnel excavation approaches a given cross-section, the soil deforms radially toward the tunnel and axially toward the working face. By the time the working face has reached the cross-section an average radial deformation, of magnitude u_1 has already occurred. If the tunnel lining was placed in contact with the soil at this point in time and was capable of preventing any further deformation of the soil mass, the average stress in the lining would be B as indicated in the figure. If further inward deformation of the tunnel walls occurred before the lining was placed, say of magnitude u_2 illustrated in the figure, the radial stress would be C.

In reality, the tunnel lining will itself undergo a radial deformation of small magnitude before stability is obtained. The effect of deflection of the lining may be estimated by a curve of its force-displacement behavior, which can be called a support reaction curve, such as the curve F in the figure. The final load

on the tunnel lining is given by the intersection of the ground reaction curve and the support reaction curve taking cognizance of the fact that a certain amount of deformation of the tunnel walls has occurred before the installation of the tunnel lining. The final stress in the tunnel lining is thus C and the deflection of the lining is u_{ℓ} . Note that the deflection of the tunnel wall is actually given by the sum $u_{1} + u_{2} + u_{\ell}$.



Average Radial Displacement Figure 5.1 Interaction of soil and tunnel lining (after Deere et al., 1969).

5.2.2 Tunnel support design concepts

The dimensioning of tunnel supports, as with any structure, requires a fairly accurate knowledge of the magnitude of the loads to be resisted by the supports. From an economics viewpoint, it is preferable to be able to estimate support requirements on the basis of exploratory drilling footage but it is certainly acceptable to be able to modify the support design based upon observations at the working face. The fact that tunnel designers have been unsuccessful in using the first method probably explains the present trend toward instrumentation of underground construction.

This is not meant to imply that there has been a lack of proposed analytic models to explain observed rock pressure and displacement; rather the major problem with the analytic models is that they lack portability. A truly general design method would have to include all possible factors such as, mass condition, material type, construction method and type of reinforcement. Since the full implications of the many factors involved, and particularly their interactions, are not presently understood, analytical techniques are typically confined to examination of a single one of the factors. This is precisely why there are no comprehensive tunnel design-load specifications

anywhere in the world and why they are compiled for each particular project on the basis of prevalent conditions.

The particular factor which is of interest in this study is the rock load for which the tunnel supports should be designed. The methods commonly in use at the present time to determine the V-8

rock pressure in the vicinity of underground excavations typically possess the characteristics of one of three categories: approximate methods based upon the extent of upbreak; theories based upon theoretical stress conditions in the rock mass; and theories based upon displacement and equilibrium assumptions. The methods which directly incorporate the jointing of the rock mass tend to be empirical rather than analytical and typically are based upon or related to the amount of upbreak above the excavation. The following brief survey of tunnel support design methods for jointed masses thus emphasizes those methods based upon the extent of upbreak. Several design concepts which do not directly include the jointing of the mass are also incorporated in the survey because they introduce concepts which are pertinent to the ensuing discussion.

The origin of the practice of dimensioning tunnel supports to resist a given amount of upbreak is usually attributed to Bierbaumer (1913), whose observations were based upon the failure of timber supports. Table 5.1 lists the values of roof pressure to be expected in various types of material. This table is frequently attributed to Bendel (1948) who actually attributes it to "others". The most significant aspect of Bierbaumer's observed

rock pressure values is that they are independent of width of the excavation.

A more widely known method of estimating support loads based upon expected upbreak is that of Terzaghi (1946). Terzaghi based his estimates of the intensity of rock loads on the failure of

Table 5.1 Observed support loads: Bierbaumer

	Roof Pre	ssure				
	$p_v(t/m^2)$		Temporary timber support			
Rock Material	After				Remark	
	At out- break	comple- tion of drift	Mode of execution	Degree of stressing		
Rock, more or less blocky	0	8-12	Skeleton lagging, light	O to in- significant	Loosening pressure small	
Very seamy rock, cemented conglomerate, soft rock, with small overburden height	10	30-35	Skeleton lagging, solid	Sma11	Loosening pressure increasing at the moment of outbreak not perceivable	
Heavily fractured rock (roof breakdown), rolling gravel and conglomerate	15-25	30-40	Tight, strong lagging	Mean	Bigger pressures perceivable simultaneously with outbreak. Ensuing of equilibrium condition, very prolongated	
Loose rock under heavy pressure (eventually in saturated condition). Bigger overburden height	25-35	40-60	Very tight, solid	Con- siderable	Stabilization of pressure conditions very difficult	
Loose and soft (pseudo- solid) rock under heavy pressure. Very big overburden height	40-60	100-150	Very tight, lagging and strong hard-wood sill-beams	Going up to rupture	Stabilization possible only after the completion of very protracted deformations (months even years; Karawanken tunnel)	

Table 5.2 Rock load guidelines: Terzaghi

Rock load H_p in feet of rock on roof of support in tunnel with width B (ft) and height H_t (ft) at depth of more than 1.5 (B+H_t)

Rock Load Hp in feet

1.	Hard and intact	zero	Light lining, required only if spalling
2.	Hard stratified or schistose	0 to 0.58	Light support.
3.	Nassive, moderately jointed	0 to 0.258	Load may change erratically from point to point.
4.	Moderately blocky	0.25B to 0.35 (B+H _t)	No side pressure.

- and seamy
- Very blocky and (0.35 to 1.10) (B+Ht) seamy
- Completely crushed 1.10 (B+Ht) but chemically intact
- Squeezing rock. (1.10 to 2.10) (B+Ht) moderate depth
- Squeezing rock. (2.10 to 4.50) (B+H_t) great depth
- 9. Swelling rock

Rock Condition

Up to 250 ft. irrespective of value of $(B^{+11}t)$ Little or no side pressure.

Considerable side pressure. Softening effect of seepage towards bottom of tunnel requires either continuous support for lower ends of ribs or circular ribs.

Remarks

Heavy side pressure, invert struts required. Circular ribs are recommended.

Circular ribs required. In extreme cases use yielding support.

wooden blocks of known strength inserted between the individual members of timber sets. The Terzaghi load estimates are summarized in Table 5.2. Note that the magnitude of the loads are dependent upon the tunnel dimensions as well as the presence or absence of groundwater.

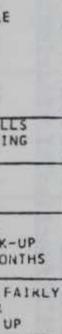
Stini (1950) also presented estimates of the rock load due to upbreak which are presented in Table 5.3. Like Terzaghi, Stini's loads are dependent upon tunnel geometry, but whereas Terzaghi described the time lag between excavation and final load (bridge-action period) as typically of the same order of magnitude as the excavation cycle time, Stini noted that much longer time periods elapsed before full loads came on the supports.

Modifications of Terzaghi's basic classification scheme are frequently found in the literature and attest to its one time high degree of acceptance. For example, a report by the California Department of Water Resources (ENR, 1959) details cost data for 99 tunnels designed by a slightly modified version of Terzaghi's basic design loads.

A major effort to add a quantifying descriptor to Terzaghi's rock load classification is due to Deere et al. (1969) and Deere et al. (1970). The pertinent data from Deere et al. (1969) is

summarized in Table 5.3. An easily measured field index properly, R.Q.D. is correlated to both Terzaghi's and Stini's classification scheme. This correlation provided the means to "objectively" select the proper load class.

SPAC		TERZAGHI(194 CLASS	6)	ROCK L	OAD Hp FINAL	REMARKS	STINI (1950) CLASS	ROCK LOAD H _p METERS	REMARKS
NI-LJ Z	86 ROD	1 HARD AN INTACT	ID	0	0	LINING ONLY IF SPALLING OR POPPING	1 STABLE	0.25+.05 B	VERY LITTLE
1'	95	HARD STRATIFIED OR	1	0	0.25 B	SPALLING COMMON		3	
		SHISTOSE 3 MAS- SIVE	56	0	0.5 B	SIDE PRESSURE IF	2 NEARLY STABLE	0.50+.10 B	FEW ROCK FALL FROM LOOSENING WITH TIME
	90_		SIVE			STRATA INCLINED, SOME SPALLING	3 LIGHTLY BROKEN	1.0 +.20 B	LCOSENING WITH TIME
6"	75	4 MODERATEL BLOCKY AN SEAMY	24 M 2	o	0.25 B TO 0.35 C		4 MEDIUM BROKEN	2.0 +.40 B	IMMEDIATELY STABLE, BREAK-U AFTER FEW MONT
_4"	50	5 VERY BLOCKY AND SEAMY, AND SHATTERED	,	0 10 0.6 C	0.35 C TO 1.1 C	LITTLE OR NO SIDE PRESSURE	5 BROKEN	5.0 +1.0 B	IMMEDIATELY FA STABLE, LATER RAPID BREAK UP
2"	25	8		area an	3.1 C	CONSIDERABLE SIDE PRESSURE. IF SEEPAGE CONTINUOUS SUPPORT	6 VERY BROKEN	7.5 +1.5 B	LOOSENS DURING EXCAVATION,LOO ROOF FALLS
	ROD IN	7 GRAVEL AND SAND	AND DESCRIPTION	0.54 C TO 1.2 C	0.62 C TO 1.38 C	classes 4-7 when und water level	AFTER DEERE ET AL.,(1969) B is tunnel width, C is width + height of tunnel		3
				0.94 C TO 1.2 C	1.08 C TO 1.38 C	For rock clar above ground reduce loads			



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Table 5.3 Rock Loads and Classification

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The effect of jointing and faulting on tunnel support loads was emphasized by Cording et al. (1971) and Cording and Deere (1972). They noted that triangular wedges could form above the crown due to adverse joint orientation and attempted to calculate the required support pressure as a function of shearing resistance along the sides of the wedge. Later work by Cording and Mahar (1974) noted that the kinematics of the situation dictated that at least one surface of the wedge should separate from the rock mass. The equivalent rock loads they presented, which are summarized in Table 5.4, do not assume any shearing resistance in the mass but are simply the pressure due to the total weight of the wedge.

The practice of designing tunnel supports on the basis of the amount of upbreak assumes that the rock has no inherent strength and that there is no real interaction between the support and the failing mass. One recent trend in tunnel support design focuses on methods which take advantage of the strength of the mass and which incorporate mass/support interaction. The brief survey of recent work is presented only to enumerate these concepts.

The "New Austrian Tunnelling Method" described by Rabcewicz (1964) is a relatively recent construction technique for minimizing the loads on tunnel supports. In the method, a thin layer of

shotcrete is applied to the tunnel walls as soon as is possible following excavation in order to prevent degradation of the rock mass and thus maintain its strength. However, as Wagner (1970) has noted, the proper use of the method requires detailed knowledge of 1000

Table 5.4 Rock loads due to crown wedges

(n8) (--) (0) HEIGHT of DIP HALF EQUIVALENT MINIMUM CONDITION ROCK LOAD ANGLE ANGLE FOR FAILURE 0 - 30 90- 60 (0 - .15)8 Both planes wavy, offset One plane wavy or offset, 30 - 45 60 - 45 (.15 · .25)B One plane smooth to slightly wavy One plane sheared, continu-45 . 60 45-30 (.25 - .45)8 258 ous and planar. One plane slightly wavy

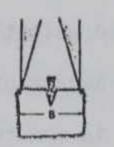
60° - 75° 30° - 15° (.45 - 1.0)8 Both planes sheared, continuous and planar

75" - 90 15" - 0" > 1.0B

15 - 0 → 1.08 sibly ure of low r

Law lateral stresses in arch; Surfaces planar, smooth, possibly open, or progressive failure aided by separation along low angle joints

From Cording and Mahar (1974)



2

the rock properties and behavior.

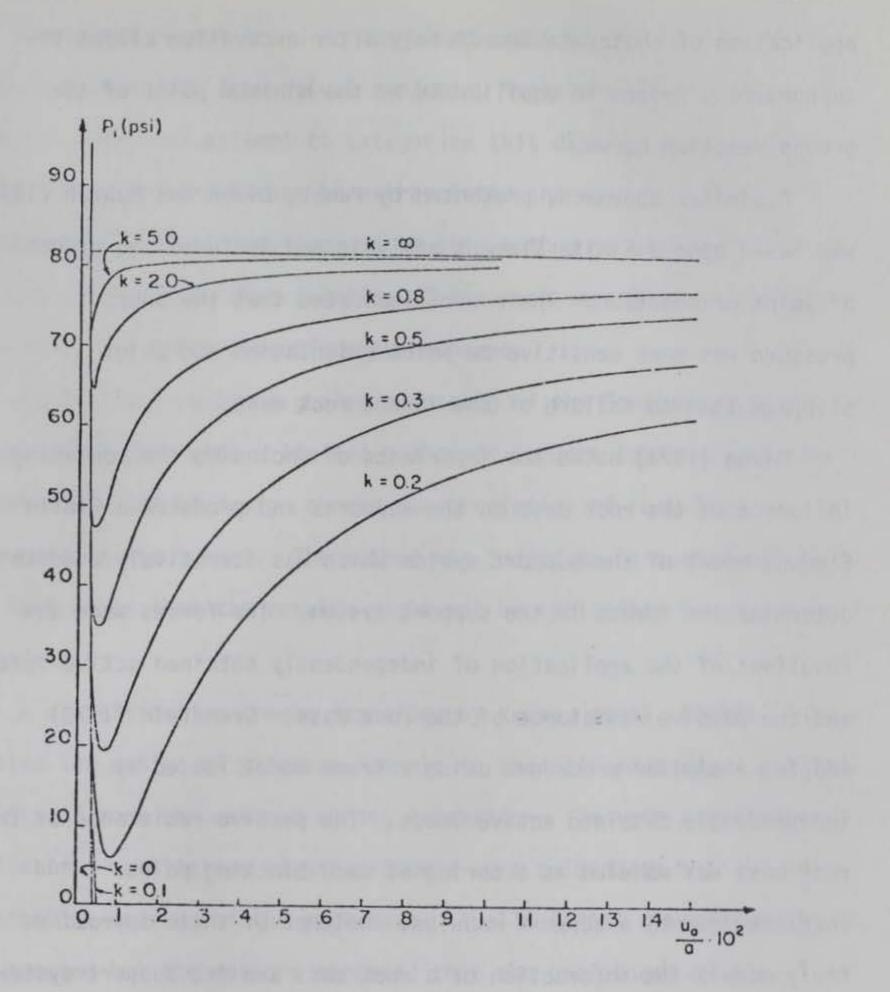
Daemen, Fairhurst and Starfield (1969), Daemen and Fairhurst (1973) and Daemen (1977) stress the need to consider both the complete force/deformation behavior of the rock mass and the interaction of the support system with the surrounding rock mass. Daemen (1977) presents ground reaction curves based upon a continuum analysis of an excavation surrounded by a zone of broken material possessing a residual strength. The method employed involved the determination of the pressure to be applied against the excavation surface to achieve stability; one resultant curve, typifying a material with low residual strength, is presented in Figure 5.2. This figure contains several interesting features. The line labeled k = . represents a material characterized by a sudden loss of strength after the peak strength is reached; note that the implication of this type of behavior is that support pressure is independent of mass deformation. This is analagous to the "dead weight" loading characteristic of the design methods based upon amount of upbreak. A second interesting feature of the figure is the two lines, labeled k = 0 and k = 0.1, corresponding to materials exhibiting perfectly plastic post peak behavior. The implication of this type of behavior is that the ground will stand unsupported; in a 15 foot diameter tunnel the strain at the

V-14

cessation of deformation corresponds to a displacement of approximately 0.1 inches.

-pproximatery off menes.

Finally, the shape of the intermediate curves lends analytical support to the practice of placing the supports early. The



Note: The parameter "k" describes post peak behavior. k = 0 is a plastic post peak behavior while $k = \infty$ is an immediate drop to a residual strength in the post peak region.

V-15

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Figure 5.2 Ground reaction curves from continuum analysis of rock with low residual strength (Daemen, 1977).

application of shotcrete immediately after excavation allows the support/mass system to equilibrate at the minimum point of the ground reaction curve.

A similar approach, presented by Panek, Dixon and Mahtab (1975), was based upon a Finite Element analysis and included the effect of joint orientation. Their work indicated that the support pressure was more sensitive to joint orientation and joint slippage than to failure of the intact rock mass.

Dixon (1971) noted the importance of including the confining influence of the rock mass on the supports and produced a Finite Element model of the support system which was iteratively used to determine the forces in the support system. The forces were the resultant of the application of independently obtained active loads and the passive resistance of the rock mass. Orenstein (1973) adopted a similar procedure using a frame model loaded by independently obtained active loads. The passive resistance of the rock mass was modeled as a spring at each blocking point characterized by a support modulus. Neither of these approaches truly models the interaction of a rock mass and its support system since the input parameters are determined independently. Typical of the methods that do model the interaction of the mass and support is that of Daemen (1975). With this model Daemen studied

the progressive development of failing material surrounding an excavation and effects of support variation. His conclusions, however, stress the need for instrumentation programs to verify this type of calculation.

The other recent trend in tunneling practice has been to collect design data from actual projects, isolate common features of the design, and attempt to categorize this data by statistical manipulation so that it can be extrapolated and used for design of new projects. The attractiveness of this method in terms of the present study is that jointing of the rock mass plays a central role in all of these classification schemes.

Abel (1966) combined geologic mapping of the Straight Creek tunnel pilot bore with a limited number of support load measurements to produce a set of design charts for prediction of rock load elsewhere in the tunnel. The method was judged to be successful but Abel noted that the results might not be applicable in other locations.

A classification scheme described by Kruse, et al. (1970) related the design of pressure tunnels to the different types and quality of rock encountered during excavation. In this particular application qualitative visual criteria were related to the deformation modulus of the rock mass. Abel's (1966) classification was adopted but the authors stressed that the usefulness of a classification scheme depended upon unambiguous definition of the input parameters.

Wickham, Tiedemann and Skinner (1972, 1974), Bieniawski (1973), and Barton, Lien and Lunde (1974) present conceptually similar classification schemes for aid in the selection of tunnel supports. The classification systems are based upon (respectively): general area geology, joint orientation and spacing, and ground water and joint condition; RQD, weathering, strength, joint spacing and orientation, joint separation, joint continuity, and ground water; and, RQD, number of joint sets, joint roughness and alteration, ground water and adverse stress conditions. All of the classification systems are relatively simple to use, utilizing data that should be routinely collected during pre-construction investigations. The methods give similar answers and can, in fact be correllated to one another (Bieniawski, 1976).

At this time it is prudent to summarize briefly those portions of the preceeding discussion which are particularly significant with respect to the present study. The majority of the methods commonly used to design support systems in jointed rock are based upon the observation of isolated failures and the extrapolation of successfully designed support systems. There is certainly nothing wrong with extrapolating previous design data to proposed ventures provided that the basic behavior mechanisms of the rock mass and support system are similar. The most significant objections to this approach are that overly conservative designs could easily propagate and that extrapolation requires a complete understanding of the pertinent geologic properties, the mass behavior, and the function of the support system.

Analytic models of the rock mass and support system provide results that indicate that the interaction of the mass and support is a significant parameter relative to the final equilibrium state. It must certainly be proper to utilize a continuum approach to study a highly stressed situation where the rock mass is failing uniformly, but there is no real evidence to suggest that this particular representation is valid for lower stressed situations where the primary deformation takes place along pre-existing discontinuity planes. In fact, the continuum analyses that have incorporated jointing in the mass indicate that the support load is more sensitive to slippage along the joint planes than to the failure of the intact mass.

The present trend of extrapolation based upon qualitatively observed parameters and instrumentation provides a useful and practical approach to the problem of tunnel support design. However, the use of these classification schemes should be guided by rationally applied analytic models wherever possible. It is precisely in this context that the Distinct Element method is used in the remainder of this chapter. In particular, ground reaction curves are presented for several realistic models in an attempt to provide a guiding rationale for the continued use of the classification schemes.

5.2.3 Calculation of the potential ultimate roof loads in the

jointed mass model

The discussion presented in Chapter 4.3 introduced a simple model for the behavior of the roofs of rooms excavated in a medium where the jointing was assumed to delineate blocks of a constant aspect ratio. The orientation of the joint planes was limited to either horizontal or vertical; additionally, the jointing in the vertical direction was assumed to be discontinuous. Subject to these restrictions, it is possible to describe a particular excavation/joint configuration in terms of three geometric parameters: the true span (0); the aspect ratio of the blocks (block thickness (t) divided by block width (w)); and the height of the triangular zone (h) which delineates that material for which unrestricted movement into the excavation is kinematically possible. These geometric parameters are noted on the diagramatic section of an excavation in a jointed mass illustrated in Figure 5.3(a). The volume of material which kinematically can undergo a finite, as opposed to an infinitesimal, displacement into the excavation is outlined and indicated in the figure.

As noted in Chapter 4.3, the number of blocks (b) in the bottom row of the roof strata and height (h) of the zone of potential finite displacement are given respectively by:

$$b = 0/w$$

and

5.1

The geometric parameters of the model can also be used to determine the total weight of the material within the triangular zone of potential finite displacement. This quantity is of interest since it represents the maximum load on the support system if the downward displacement of the triangular zone is sufficient to cause loss of transmittal of vertical force across the boundary between the triangular zone and the overlaying strata.

The total weight (L) of material within the triangular zone is easily calculated in terms of the total number of blocks (B) comprising the zone. For a unit thickness normal to the plane of the paper and a given weight density (d), the total weight within the zone of potential finite displacement of the basic model illustrated in Figure 5.3(a) is:

$$= B \cdot t \cdot w \cdot d$$
 5.2

The total number of blocks within the zone of potential finite displacement is related to the true span of the excavation and the block width. In fact, it is the quotient of these two parameters, the number of blocks in the bottom row, that leads to a simple expression for the total number of blocks in the triangular zone. The total number of blocks in the triangular zone is the sum of the number of blocks in each of n rows of blocks in the zone:

 $B = b + (b-1) + \dots + (b-n+2) + (b-n+1)$ 5.3 The terms on the right side of the equal sign in equation 5.3 are the terms of an arithmetic progression

$$a_n = a_1 + (n-1) d$$
 5.4

where a₁ is the first term,

a_n is the nth term, and
d is the common difference

The properties of the basic jointed mass model are such that:

$$a_1 = b$$
,
 $a_n = 1$,
 $n = b$, and
 $d = -2$

The total number of blocks in the triangular zone is given by the

sum of the first n terms of this arithmetic progression: $B = \frac{b}{2} (b + 1)$ 5.6 The total weight of material within the zone of potential finite displacement is thus:

 $L = \frac{b}{2} (b + 1) . t . w . d$ In terms of the true span of the excavation: 5.7

$$L = \frac{0t}{2} \left(\frac{0}{w} + 1\right) d$$
 5.8

Equation 5.8 was used to obtain the five sets of curves presented in Figure 5.3. Each family of curves represents a constant block width while each curve within a family represents a different block thickness. The thickness values increase in an upward direction. The calculations were performed using a weight density of 150 pcf; all length dimensions are thus in feet. Since equation 5.8 is linear with respect to density, the curves may be corrected for any desired density simply by multiplying the load by the quotient of the desired density, in pounds per cubic foot, and 150 pcf.

The graphs illustrated in Figure 5.3 should be used with caution since the model upon which they are derived is based upon integer values of the number of blocks in the lower row. Although the curves give a seemingly proper value of the load for non-integer values of b, the jointed model is only defined for those instances where the span is an integer multiple of the block width. It must

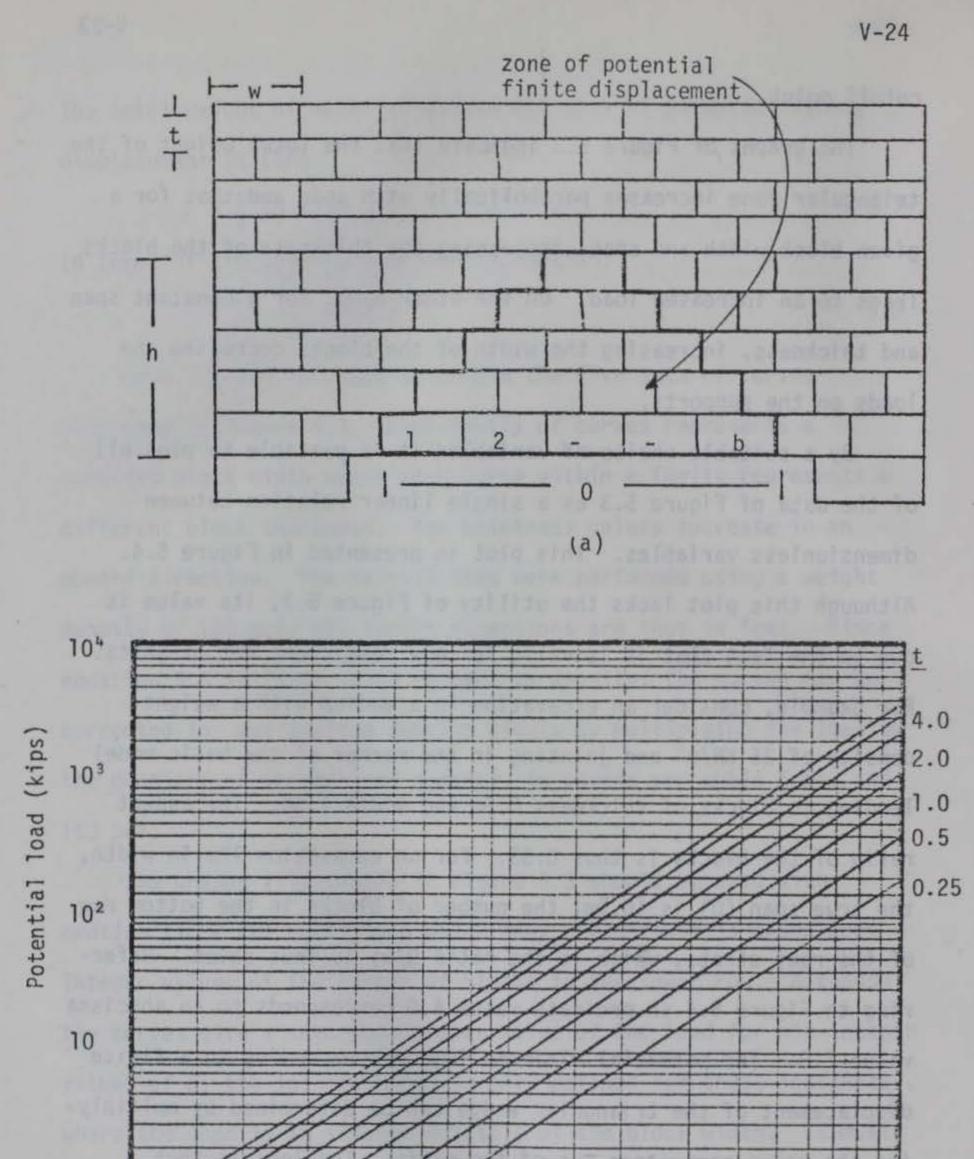
also be noted that even though the complete curves have been plotted in all cases, the model is also undefined in those instances where the true span is less than the block width. This cutoff point has been indicated on the abscissa of each plot by a small triangle; the curves are not valid for the basic model to the left of this cutoff point.

The graphs of Figure 5.3 indicate that the total weight of the triangular zone increases parabolically with span and that for a given block width and span, increasing the thickness of the blocks leads to an increased load. On the other hand, for a constant span and thickness, increasing the width of the blocks decreases the loads on the supports.

By a suitable choice of variables it is possible to plot all of the data of Figure 5.3 as a single linear relation between dimensionless variables. This plot is presented in Figure 5.4. Although this plot lacks the utility of Figure 5.3, its value is due to the fact that it is valid for any consistent set of units. For example, consider an excavation in a medium with a weight density of 26 KN/m³ and jointing in the manner of the basic model leading to blocks of thickness 0.5m and width 1.5m. The aspect ratio of the blocks is thus 0.33. For an excavation 12m in width, the true span (0) is 10.5m; the number of blocks in the bottom row of the roof strata, which is the ratio 0/w; is thus seven. Referring to Figure 5.4 an ordinate value 4.0 corresponds to an abscissa value 7.0. The potential ultimate load corresponding to a finice displacement of the triangular wedge can be determined by multiply-

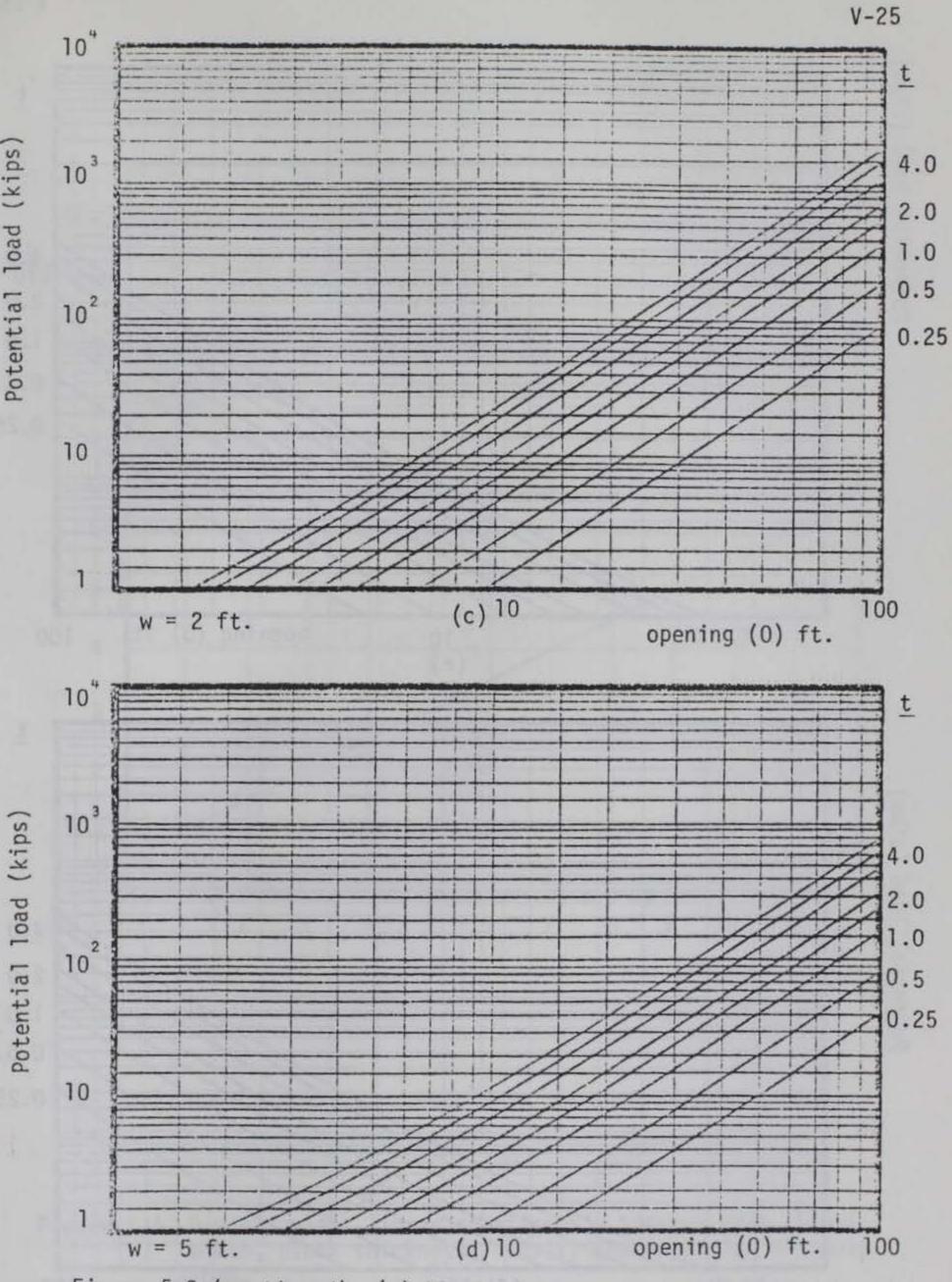
V-23

ing the known parameters out of the ratio. The load is thus $4 \times 10.5m \times 0.5m \times 26 \text{ KN/m}^3$ or 546 KN per meter of excavation length.



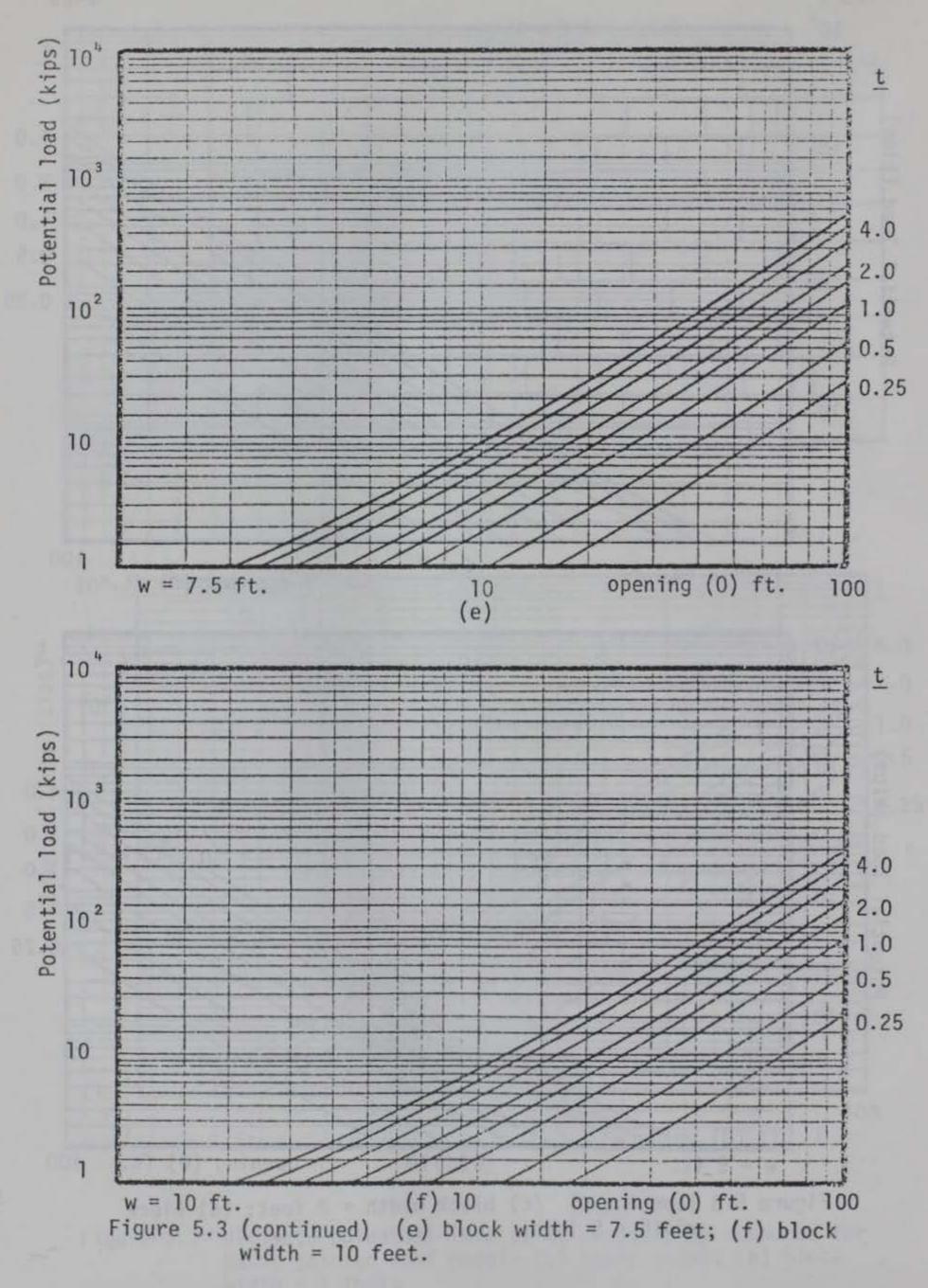
$1 \frac{1}{10} \frac{10}{100}$ w = 1 ft (b) 0pening (0) ft.

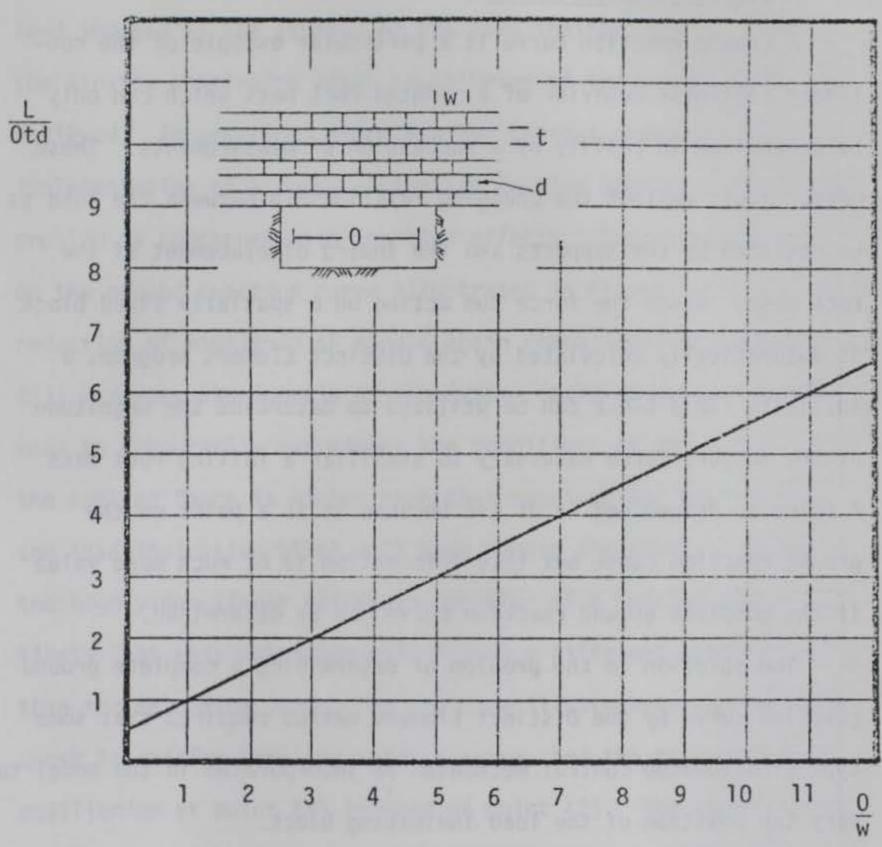
Figure 5.3 Ultimate potential load to be resisted by supports for basic jointed roof model: (a) basic model; (b) block width = 1 foot;



Potential load (kips)

Figure 5.3 (continued) (c) block width = 2 feet; (d) block width = 5 feet;





V-27

Figure 5.4 Diminsionless linear relationship between span, block width, block thickness, density and potential ultimate load.

5.2.4 The use of displacement controlled fixed blocks to generate ground reaction curves

A ground reaction curve is a particular example of the nonlinear stiffness behavior of a jointed rock mass which can only be determined in reality by a succession of measurements. These measurements reflect the changing relationship between the load to be resisted by the supports and the inward displacement of the rock mass. Since the force sum acting on a spatially fixed block is automatically calculated by the Distinct Element program, a spatially fixed block can be utilized to determine the magnitude of the support force necessary to stabilize a failing rock mass. A value so determined is of use because it is a point on the ground reaction curve but this information is of much more value if the complete ground reaction curve can be determined.

The solution to the problem of determining a complete ground reaction curve by the Distinct Element method requires that some type of automated control mechanism be incorporated in the model to vary the position of the load indicating block.

Analogous to a laboratory testing frame, there are two basic governing control mechanisms: force control, which requires a freely moving block; and displacement control which requires a

spatially fixed block. Both mechanisms require that a small block be placed against the strata in the manner illustrated in Figure 5.5(a) and (b).

To implement the force controlled testing machine, the force

on the load indicating block is reduced by some amount. The net result of this action would be an acceleration, due to the excess load imposed by the strata, of the load indicating block away from the strata, continuing until equilibrium of the system was again achieved. In practice, there are two serious drawbacks to the implementation of a force controlled testing machine. The first problem is concerned with inertial effects. Beginning at point (1) on the ground reaction curve illustrated in Figure 5.5(c), a force reduction of magnitude ΔF should again reach equilibrium at point (2); however, the inertia of the system could cause the jointed mass to temporarily experience the conditions at point (3). Since the applied force is higher than that required for equilibrium, the load indicating block will move toward the strata. Owing to the highly non-linear stiffness behavior of a jointed mass, it is likely that this reloading will follow a different behavior curve than the unloading curve. In the case illustrated, the reloading curve is stiffer than the loading curve, and the mass comes to equilibrium at point (4) instead of point (2). The result of this is that instead of the true ground reaction curve (1) - (2) - (3), the data would indicate curve (1) - (4) as being the ground reaction

curve.

V-29

The second problem that would be encountered would occur if the ground reaction curve had an upswing such as the segment of the curve (6) - (7) in Figure 5.5(c). The postulated force controlled testing machine would continue to lower the force applied

to the load indicating block and thus, equilibrium could not be reached.

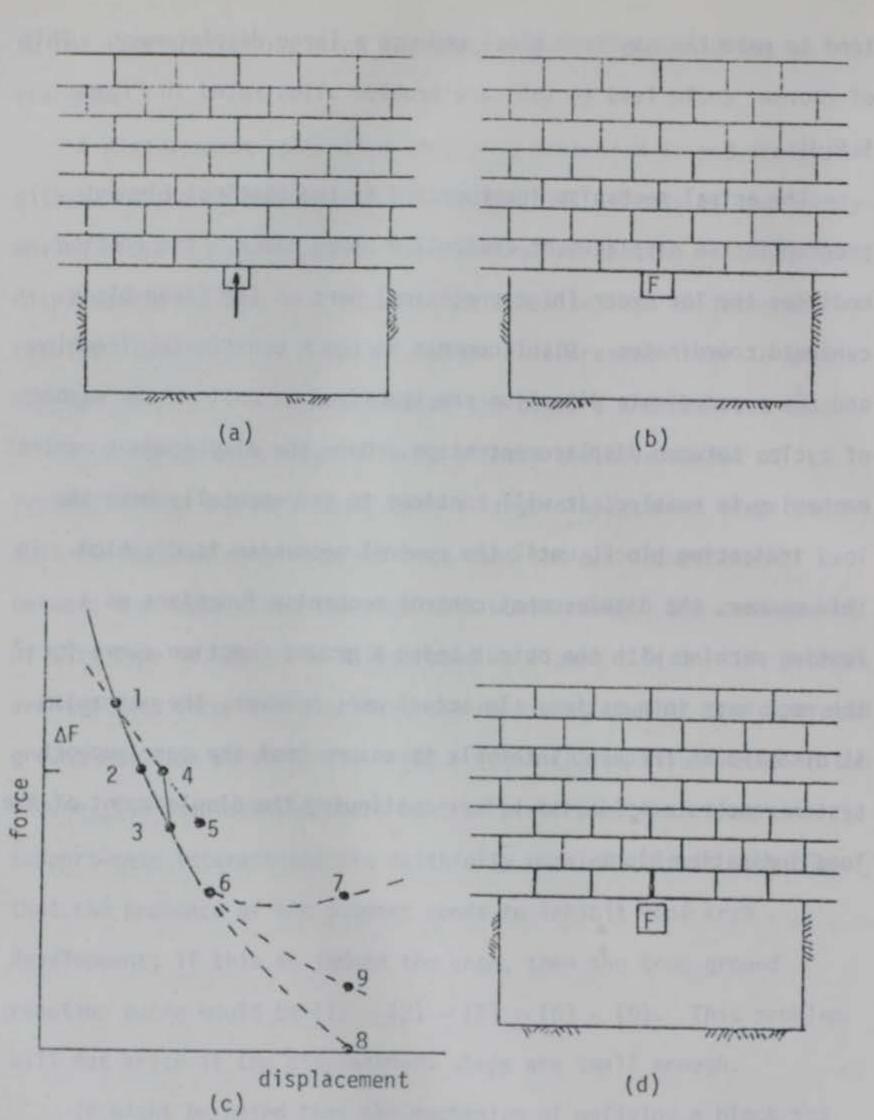
A displacement controlled governing mechanism is not foolproof either. Although not subject to the inertial effects of the freely moving block utilized in the force controlled testing machine, the displacement control of a fixed block can also lead to incorrect results. One point of interest, which is addressed later in this chapter concerns the interaction of the support and the rock mass. If the presence of a support force affects the development of arching within the rock mass, then a large displacement step could pull the support away from the rock mass and all interaction between the support and the rock mass would cease. One consequence of this type of action is illustrated in Figure 5.5(d). If, indeed, arching does occur and stabilize the rock mass so that the generated ground reaction curve is (1) - (2) - (3) - (8) as illustrated in the figure, the displacement steps must be small enough so that the support-mass interactions are faithfully modeled. It is possible that the presence of the support tends to inhibit roof arch development; if this is indeed the case, then the true ground reaction curve would be (1) - (2) - (3) - (6) - (9). This problem will not arise if the displacement steps are small enough.

It might be noted that the mechanism of unfixing a block and

letting it move to a new position before refixing it does not lead to an acceptable solution. The force sum acting on the fixed block is a large quantity relative to the weight of the fixed block. Thus when the fixity of the block is removed, high acceleration would

tend to make the now free block undergo a large displacement. This of course, could lead to the same problem illustrated in Figure 5.5(d).

The actual mechanism incorporated in the Distinct Element program is the displacement controlled fixed block. The routine modifies the low order (high precision) part of the fixed block centroid coordinates. Displacements in the x coordinate direction and the y coordinate direction are specified as well as the number of cycles between displacement steps. Once the displacement control mechanism is enabled, it will continue to incrementally move the load indicating block, until the control mechanism is disabled. In this manner, the displacement control mechanism functions as a testing machine with the output being a ground reaction curve for the rock mass in question. In actual use, however, the mechanism is disabled at frequent intervals to ensure that the mass/support system reaches equilibrium before continuing the displacement of the load indicating block.



Mechanisms for obtaining ground reaction curves for jointed rock mass (a, b and d) and generalized force displacement curve (c). Figure 5.5

5.3 Support Requirements in the Absence of Arch Development

In order that the development of the ideas presented in this chapter be complete, it is prudent to examine the support requirements for the simple monolithic roof model presented in Chapter 4.4. Recall that owing to the absence of flexural deformation in the model, arching behavior was unable to develop and stability of the single block was achieved by frictional resistance acting along the vertical joints. For those situations where the magnitude of the horizontal force acting on the block is insufficient to prevent failure of the roof through downward movement of the block, equilibrium, and thus the integrety of the roof, can only be obtained by the application of an external force.

The Limit Equilibrium models utilized in Chapter 4 can easily be modified to incorporate an external force or the resultant of an external support pressure; the modified models are illustrated in Figure 5.6(a). The assumptions of symmetry of the frictional reactions and the full mobilization of frictional resistance lead to an equation of vertical equilibrium which is given by:

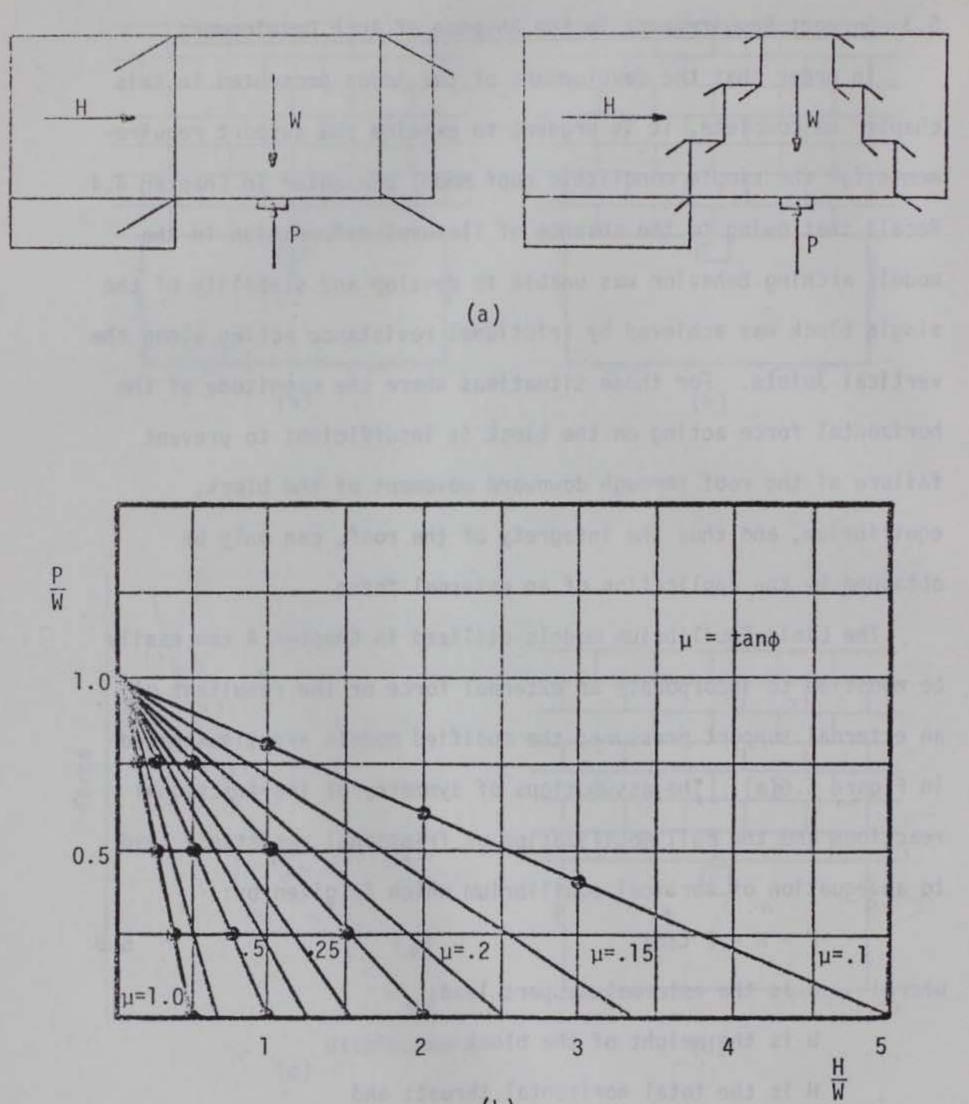
 $P = W - 2 \tan \phi$

where: P is the external support load;

W is the weight of the block

H is the total horizontal thrust; and \$\phi\$ is the angle of sliding friction of the joints.
If the support load and horizontal thrust are normalized with
respect to the weight, a diminsionless form of equation 5.9, V-33

5.9



(b)

Figure 5.6 (a) Limit Equilibrium models of roof behavior under combined frictional suspension and external force. (b) external support requirement for stability of frictionally suspended roofs.

$$\frac{P}{W} = 1 - \frac{2H}{W} \tan \phi$$
 5.10

is obtained. This equation is plotted in Figure 5.6(b) for various values of tand. As was expected, the magnitude of the external support force decreases with increasing horizontal thrust; the decrease is more rapid for higher joint friction angles.

A number of unstable, monolithic roof geometries were modeled using the Distinct Element method for purposes of comparison to equation 5.10. In these models the external support load required for stability was either applied to the centroid of the roof block or applied to the centroid of a small block placed at midspan on the bottom of the roof block specifically for this purpose. There was no discernable difference in the results obtained by the different methods. Examination of Figure 5.6(b) reveals a high degree of correlation between the Limit Equilibrium solution and those calculated by the Distinct Element method.

The basic model dealt with in this study forms an inverted "staircase" in the roof when failure occurs (see Chapter 4.3). The geometric relationships relating total roof load to the span of the excavation and the aspect ratio of the blocks formed by the jointing which were developed in the preceeding section can be used to determine the magnitude of the parameter W in equation 5.9. Bearing

V-35

in mind the fact that the roof is monolithic it is still possible to calculate a ficticious aspect ratio for the joints that form the vertical sides of the roof block. Thus equation 5.7 or 5.8 may be used to determine the total weight of the roof. If the support

force is assumed to be some percentage (K) of the total roof load and if in addition, the total horizontal thrust (H) is expressed as the height of the arch (h) multiplied by the horizontal stress (σ_h) , then K is given by the relation:

$$K = 1 - 2 \frac{0 \frac{t}{w} \sigma_{h} \tan \phi}{(\frac{0^{2} t}{2w} + \frac{0t}{2}) d}$$
 5.11(b)

K = 1 - 4R/(0 + w) 5.12

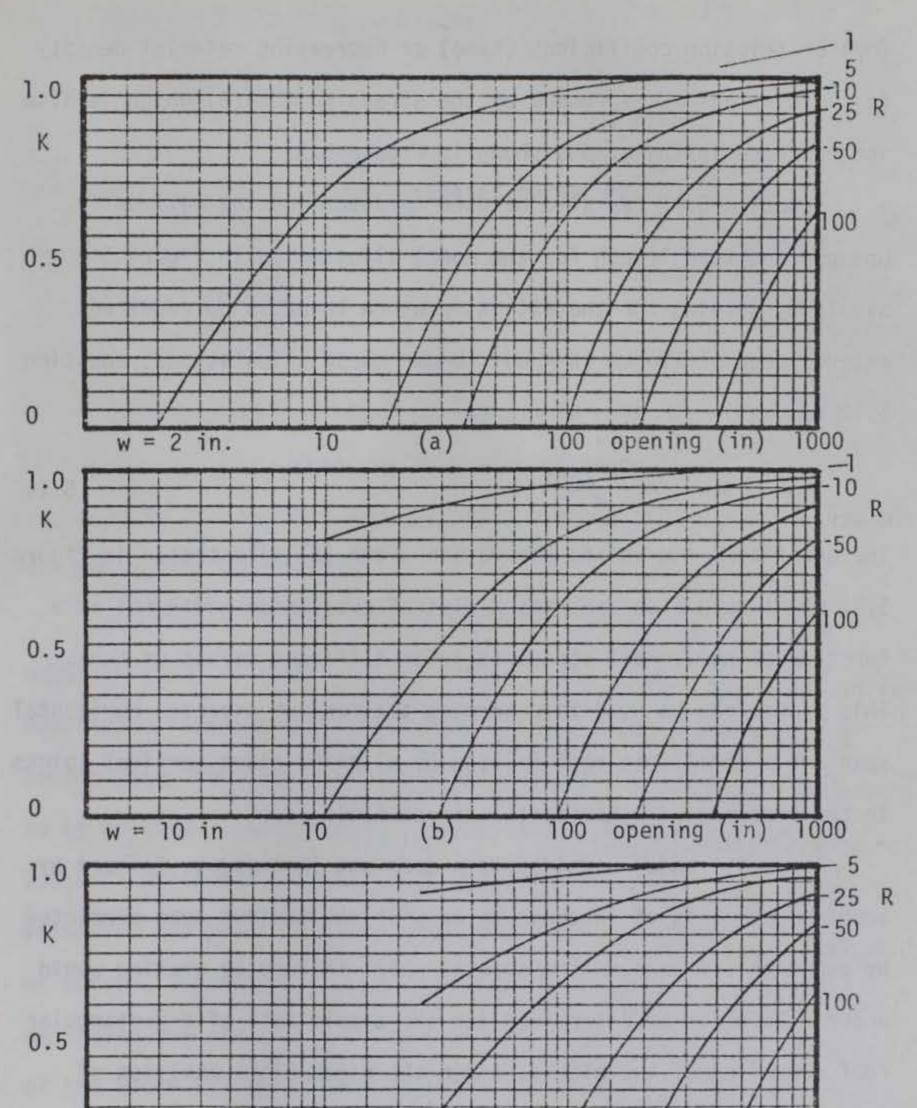
The stress factor (R) is defined as

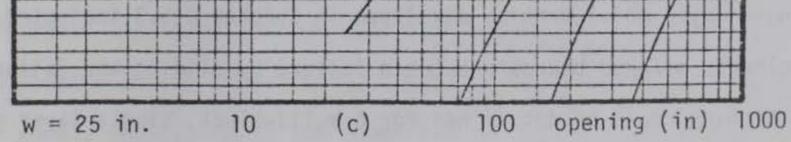
$$R = \frac{\sigma_{h} \tan \phi}{d}$$
 5.13

All of the above mentioned parameters are illustrated in Figure 5.8.

Figure 5.7 illustrates the relationship between the percentage of the roof load to be supported (K), the true opening width (O), the stress factor (R) and the block width (w). The three separate graphs correspond to different values of w, chosen to represent: a high fracture frequency or a low RQD (w = 2 in.); a moderate fracture frequency or RQD (w = 10 in.) and; a low fracture frequency or a high RQD (w = 25 in.). The curves demonstrate an increase in the percentage of support required corresponding to an increase in

block width; this reflects the fact that for any given block thickness, an increase in the block width tends to make the roof block assume a rectangular rather than a triangular shape. The percentage of support required also decreases with increasing horizontal stress





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Figure 5.7 Percentage of total roof weight (k) to be supported as a function of true opening (0) for varying block width (w) and stress factor (R).

 (σ_h) or friction coefficient (tan ϕ) or decreasing material density (d). This fact is expressed by the stress factor (R) which is also incorporated in the graphs shown in Figure 5.7.

Equation 5.12 can also be used to determine the maximum unsupported span length for the model illustrated in Figure 5.8 simply by solving for the situation where there is no required external support force (K = 0). Under these stipulations, equation 5.12 becomes:

$$0 + w = 4 \frac{\sigma_h \tan \phi}{d}$$
 5.14

The quantity 0 + w is the excavation width (S) illustrated in Figure 5.8; the figure also presents a plot of excavation width (S) as a function of horizontal stress (σ_h) for different values of tan ϕ . This figure can be used to determine the maximum expected horizontal span for a monolithic roof failing by slipping along vertical joints in the presence of a horizontal stress field.

The model under consideration does not incorporate failure by arching but it is of interest to know if the maximum span predicted by equation 5.6 exceeds the span at which failure by arching would occur. This can be determined for the simple case of a rectangular roof comprised of two blocks, since the rigid block analyses of

single layer model arching developed in Chapter 4.5.3 indicated that a clearly defined boundary between failure by sliding and failure by arching could be determined for a multi-block, single layer model. In terms of maximum unsupported spans for a two block rectangular roof, equation 4.3 may be rewritten:

$$0 = 2 \frac{\sigma_h}{d} \tan \phi \qquad 5.15$$

Likewise, equation 4.9, which relates horizontal thrust to span may be rewritten:

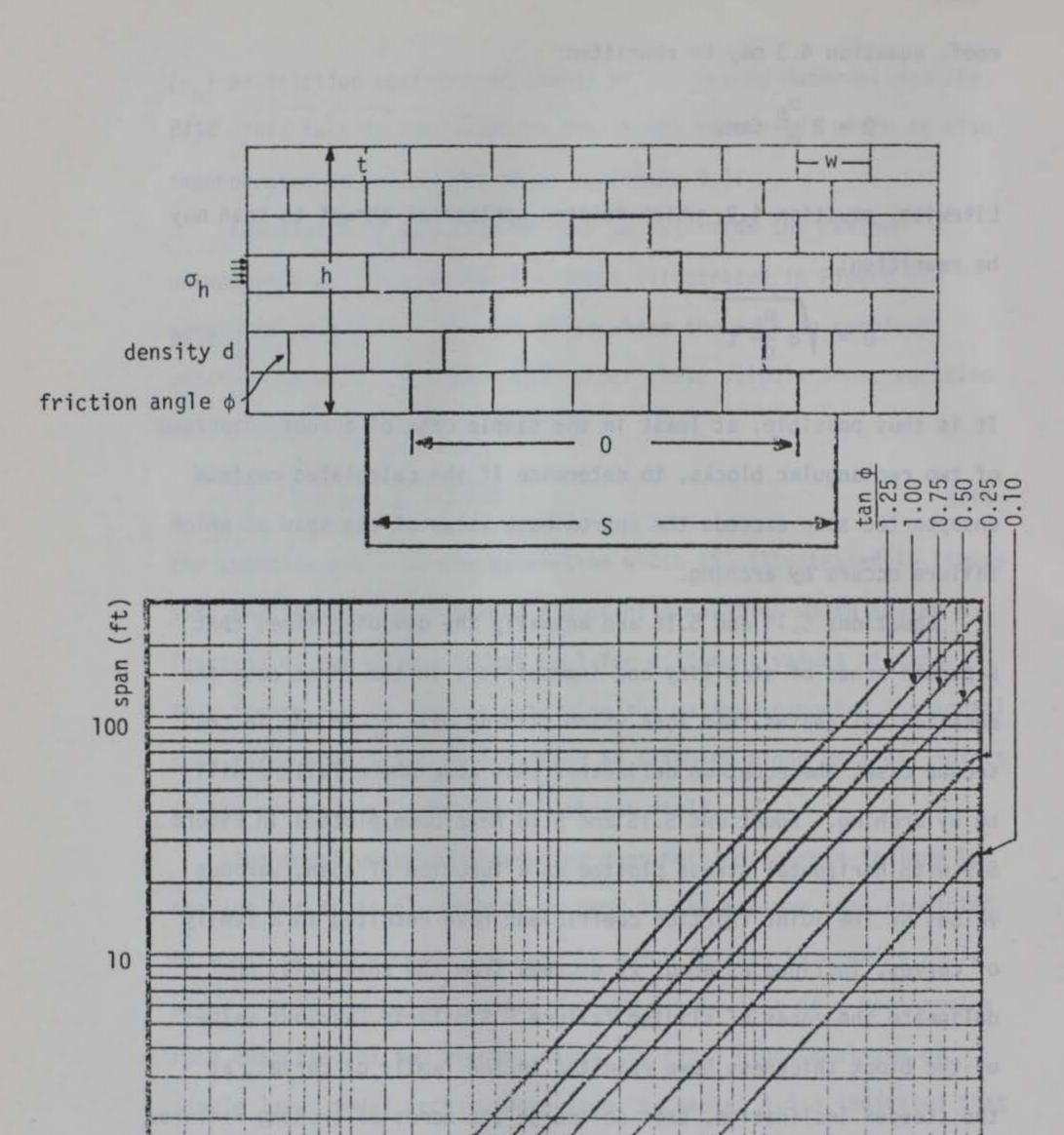
$$0 = \sqrt{8} \frac{\sigma_h}{d} t$$
 5.16

It is thus possible, at least in the simple case of a roof comprised of two rectangular blocks, to determine if the calculated maximum unsupported span exceeds the approximate value of the span at which failure occurs by arching.

Equations 5.15 and 5.16 are actually the dividing lines that separate zones of stability and instability; in the first case the equation delineates that zone where sliding will occur and in the second case, the equation delineates that zone where failure will be by arching. Equations 5.15 and 5.16 have been plotted in Figure 5.9 with horizontal stress plotted as a function of span, various values of the joint friction coefficient have resulted in a family of curves, inclined at about 25 degrees from the span axis, that delineate the zones of sliding failure. Similarly, various values of the block thickness have resulted in the family of curves, at

the steeper inclination, that delineate the zones of arching failure. When plotted on the same figure, these two equations thus delineate four zones, indicative of the condition of the roof, that are

dependent upon the block thickness and the joint friction

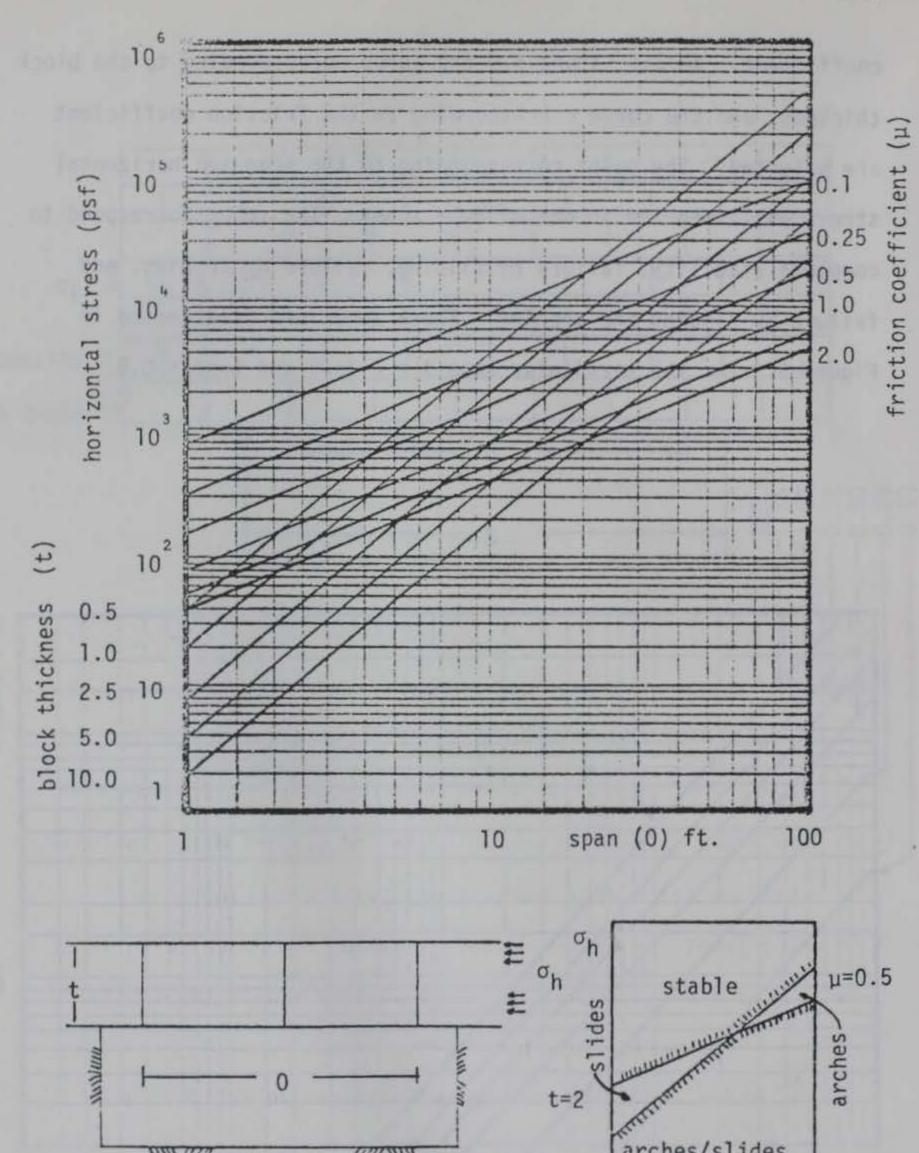


1

coefficient. To use Figure 5.9 the curve corresponding to the block thickness and the curve corresponding to the friction coefficient are selected. The point corresponding to the span and horizontal stress will then lie in one of four zones. The zones correspond to complete stability, failure by sliding, failure by arching, and failure by sliding and arching. These zones are illustrated in Figure 5.9 for the particular case t = 2 feet and $tan\phi = 0.5$.

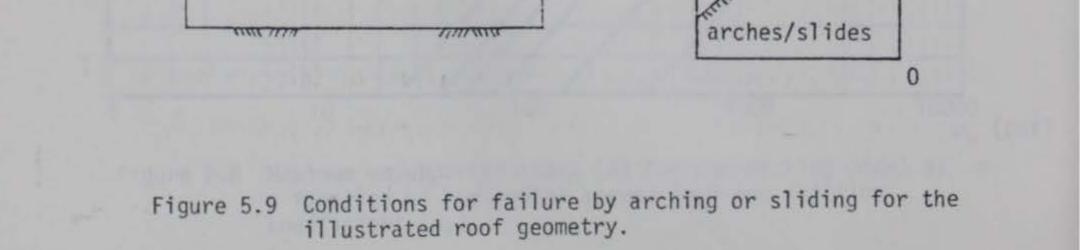
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5.4 <u>An Investigation of Support Requirements in Jointed Roofs</u>

5.4.1 Jointed mass behavior representation by means of ground reaction curves

The brief survey of design concepts presently in use to aid in the dimensioning of tunnel supports indicated that the majority of the methods that recognize the rock mass as a jointed discontinuum are of an empirical nature and are often criticized for their failure to account for the interaction of the support system and the rock mass. However, the fact that the older amount of upbreak or dead weight loading schemes (Bierbaumer, Terzaghi and Stini) are based upon observations, admittedly crude, of pressures acting on installed support systems indicates that there is at least some partial measure of the support/mass interaction incorporated within them. The same is true of the newer schemes (Wickman, Tiedeman and Skinner, Bieniawski, and Barton); the design pressures are based upon actual installed support data supplemented by instrumentation data where it was available. Thus the interaction of the mass and support system is incorporated in these schemes even though it is not somehow explicitly expressed as one of the basic input parameters.

Conspicuous in its absence, however, is analytical substantia-

tion of the required support loads predicted by the empirical schemes for those instances where the failure of the rock mass and the resulting loading of the support system is governed by the presence of distinct planes of weakness, such as joints and

faults, within the rock mass. The Distinct Element method provides the mechanism to investigate the behavior of jointed masses which are controlled by the behavior of the joints. Additionally, the implementation of the displacement controlled testing mechanism described in Chapter 5.2.4 provides the data necessary to quantitatively describe the behavior of the jointed rock mass as it interacts with a simple support system.

The Distinct Element method has been used to study the support requirements of numerous excavation roofs which possess the joint pattern characteristic of the basic model utilized in Chapter 4. These characteristics are regular, continuous jointing in the horizontal direction and regular, discontinuous jointing in the vertical direction. Once again, this is a plane strain model and the aspect ratio of the blocks for a given problem is a constant. The results of this investigation are presented in this section by means of several ground reaction curves which are representative of the observed responses.

The results presented in Chapter 4 indicated that the stability of the roof of an excavation in jointed rock was most sensitive to the magnitude of the horizontal stress. It follows logically, therefore, that an investigation of the support requirements of excavations in jointed media should be concerned with the effect of horizontal stress on the ground behavior as expressed by a ground reaction curve relating the total load acting on the support to the vertical deflection of the support.

The models analyzed in this chapter are subject to the limitations of those described in Chapter 4, namely highly idealized joint behavior and a simplified mechanism for modeling the horizontal stress. The joints are modeled as planar and do not possess cohesion. The tendency of construction procedures such as blasting is to destroy the cohesion of the joint surfaces near the excavation. This, coupled with the fact that the models portray the behavior of failing masses leads to the conclusion that the analyses are valid in terms of the cohesive strength of the joints. The fact that the joints are considered to be planar, however, does detract somewhat from the validity of the analyses. Real joints are non-planar; perfectly mating rough surfaces can only be forced to slide relative to one another if they are free to move apart. This dilatancy leads to increased mass strength for if the joint separates two confined blocks, the only way relative movement can occur is if shearing of the rock mass takes place. As noted in Chapter 4.5.2, the horizontal stress field is modeled as a constant load, owing to the rigid nature of the blocks in the Distinct Element formulation. Under a constant load situation strength increases due to dilatancy do not occur. The analyses presented in this chapter are probably only realistic for problems where dilatancy does not play a significant role.

V-45

Near surface excavations with relatively open or infilled jointing

are examples of such a situation.

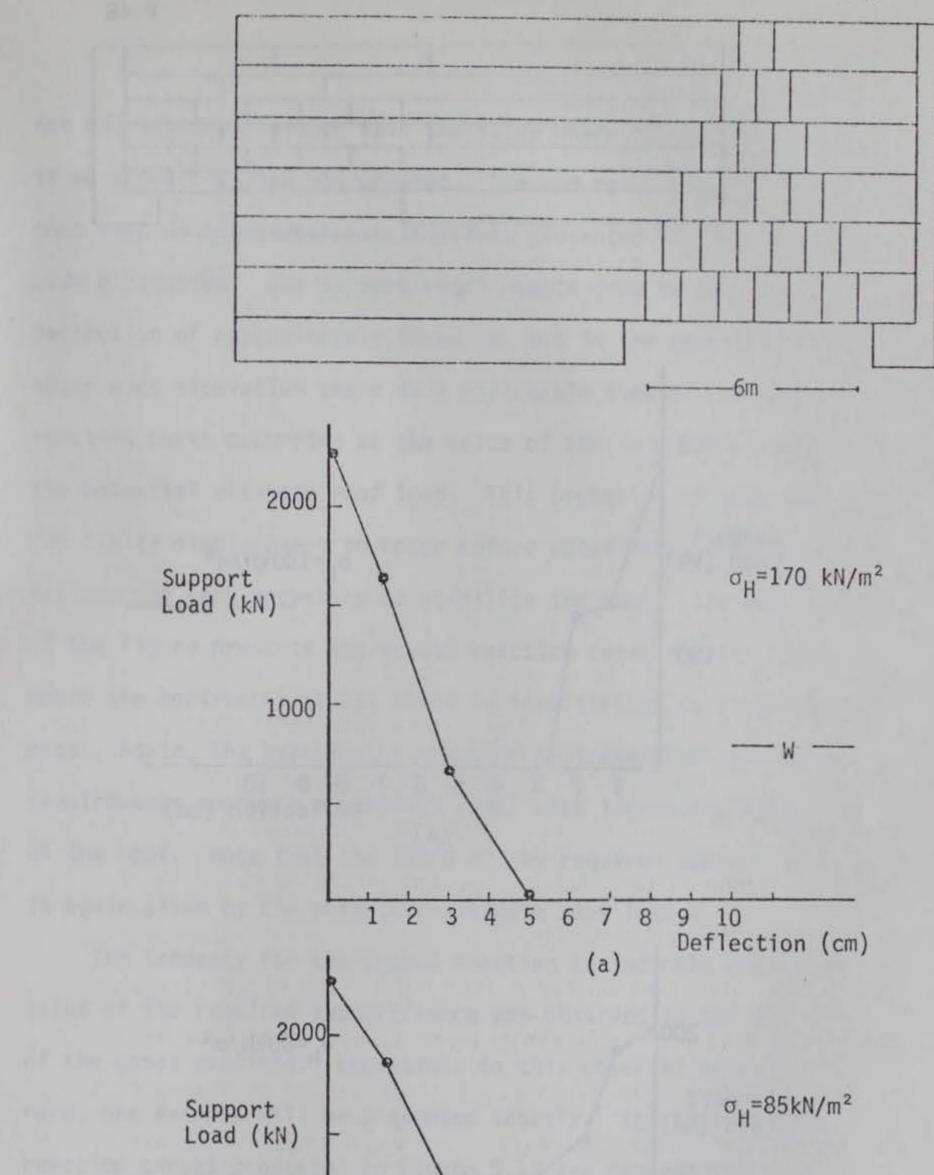
Figure 5.10 presents two ground reaction curves for the six

meter wide excavation illustrated in the figure. Part (a) of the figure illustrates the ground reaction curve for a case where sufficient horizontal stress exists to stabilize the mass in the absence of externally applied support. The ground reaction curve reflects this fact indicating that a value of the roof deflection of approximately five centimeters, the load acting on the supports is zero. The second ground reaction curve illustrated in the figure represents a situation where the magnitude of the horizontal stress field is insufficient to stabilize the mass without the introduction of external support. The parameter W, indicated on the ground reaction curve, is the total weight of the material within the zone of potential finite displacement described in Chapter 5.2.3. W is thus that quantity which was previously termed the potential ultimate roof load. The form of the ground reaction curve suggests that as deflection of the roof continues the required support force approaches a constant value, and that this value is given by the potential ultimate roof load W.

A similar situation for a four meter wide excavation where the blocks have a significantly lower aspect ratio (0.4 as opposed to 1.5 for the first case) is presented in Figure 5.11. As before, the two ground reaction curves represent the situations where sufficient stabilizing horizontal pressure is present

(part a) and the case where external support is required for stability for the roof (part b). However, in this case, the ground reaction curve in the first part of the figure represents the behavior of the mass where the applied horizontal stress is

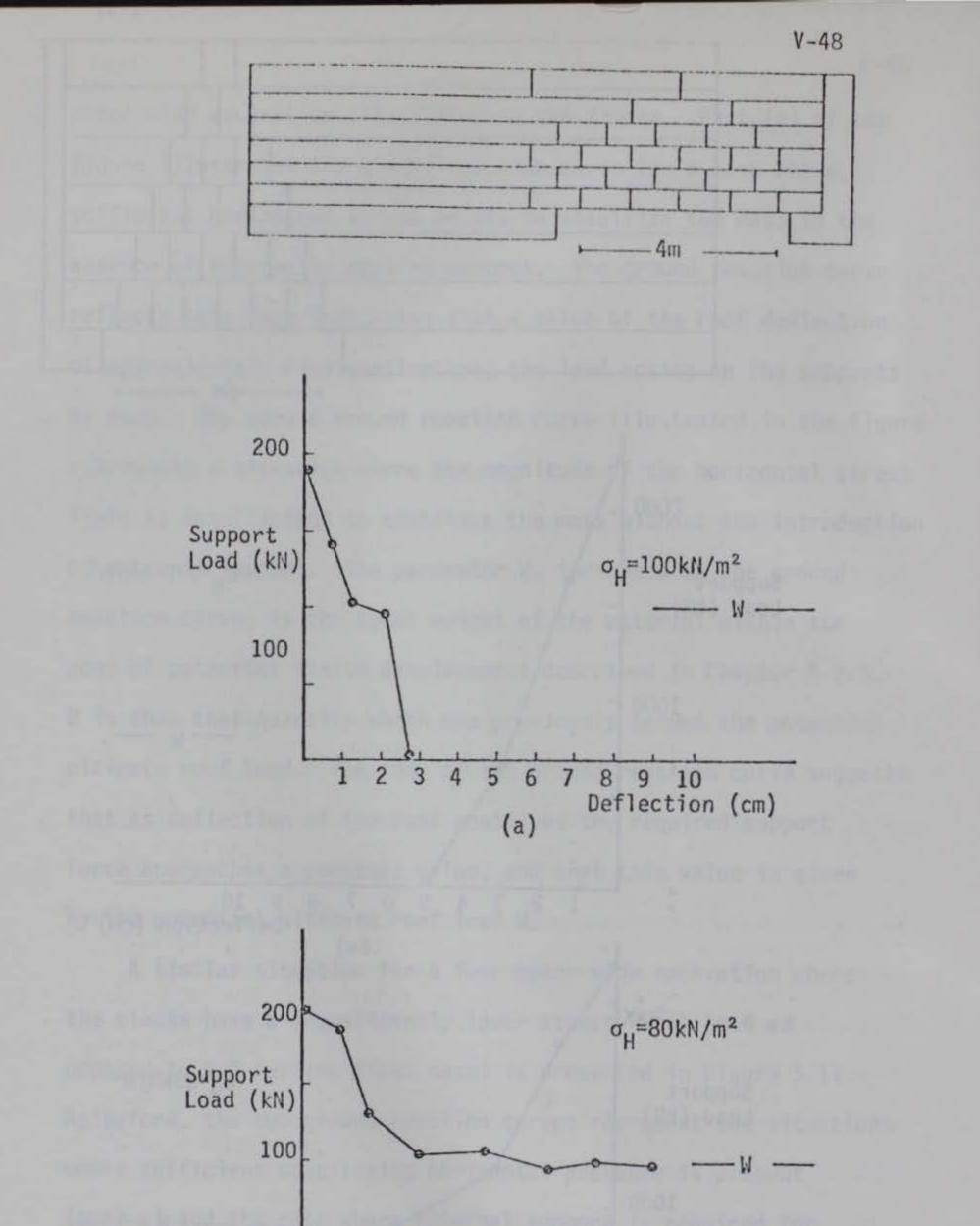




1000

(b)

1 2 3 4 5 6 7 8 9 10 Deflection (cm) Figure 5.10 Ground Reaction Curves for 6m Wide Excavation:(a) High Horizontal Stress;(b) Low Stress.



1 2 3 4 5 6 7 8 9 10 (b) Deflection (cm)

Figure 5.11 Ground Reaction Curves for 4m Wide Excavation: (a)Stabilizing Horizontal Stress; (b) Insufficient Horizontal Stabilizing Stress.

not significantly higher than the value where failure would occur if no support system was present. The end result is the same as that seen in higher stress situations presented for the six meter wide excavation. The support requirements drop to zero at a roof deflection of approximately three cm, but in the case of the four meter wide excavation there is a noticeable kink in the ground reaction curve occurring at the value of the load corresponding to the potential ultimate roof load. This probably reflects the need for finite displacement to occur before rotation of the blocks can devleop the arch necessary to stabilize the roof. The second part of the figure presents the ground reaction curve for the situation where the horizontal stress alone is insufficient to stabilize the mass. Again, the behavior of the roof indicates that the support requirements approach a constant level with increasing deflection of the roof. Note that the value of the required support resistance is again given by the potential ultimate roof load W.

The tendency for the ground reaction to indicate a constant value of the required support force was observed in the majority of the cases examined. Exceptions to this observed behavior were rare; one example will be presented shortly. The three ground reaction curves presented in Figure 5.12 are representative of a

number of calculated mass responses and indicate that the rock load for which supports should be designed is represented fairly accurately by the potential ultimate roof load. Figure 5.12(a) and (b) both represent situations of insufficient horizontal stabilizing force for a

16m 2000 $\sigma_{\rm H}$ =1500kN/m² μ=0.5 Support Load (kN) 1000 10 20

Deflection (cm) (a)

Figure 5.12 Ground Reaction Curves for a 16 meter Wide Excavation Illustrating the Consistancy of Constant Support Load with Decreasing Horizontal Stress and Friction Coefficient.

V-51 2000 Support Load (kN) $\sigma_{\rm H}^{=500\,\rm kN/m^2}$ μ=0.5 1000 W 20 Deflection (cm) 10 (b) 2000 Support Load (kN) $\sigma_{\rm H}$ =500kN/m² µ=1.0 1000 W

Deflection (cm) 10 (c) Figure 5.12 Continued.

16 meter wide excavation; part (b) however, represents a situation of much lower horizontal stress. The general shape of the ground reaction curves is, nevertheless, similar. The third ground reaction curve also represents low stress conditions but indicates the effect of increasing the friction coefficient of the joints. As can be seen, the same constant load requirement emerges. The major effect of the higher friction coefficient is to decrease the rate at which the ground reaction curve drops to the final, constant level. This is also representative of other cases observed; an increase in the friction coefficient has little effect on the ultimate support requirement.

The three curves presented in Figure 5.12 also indicate a characteristic decrease in the support load requirements with further roof deflection. This decrease in required support was observed most frequently in problems involving blocks with a low aspect ratio. This behavior typically corresponded to roof deflections of the order of 10 to 20 percent of the block thickness and is indicative of bed separation occurring as an arch develops in the second row of strata above the excavation. This behavior was not observed in situations involving higher aspect ratios, probably owing to the tendency of this type of model to fail by sliding rather than arching.

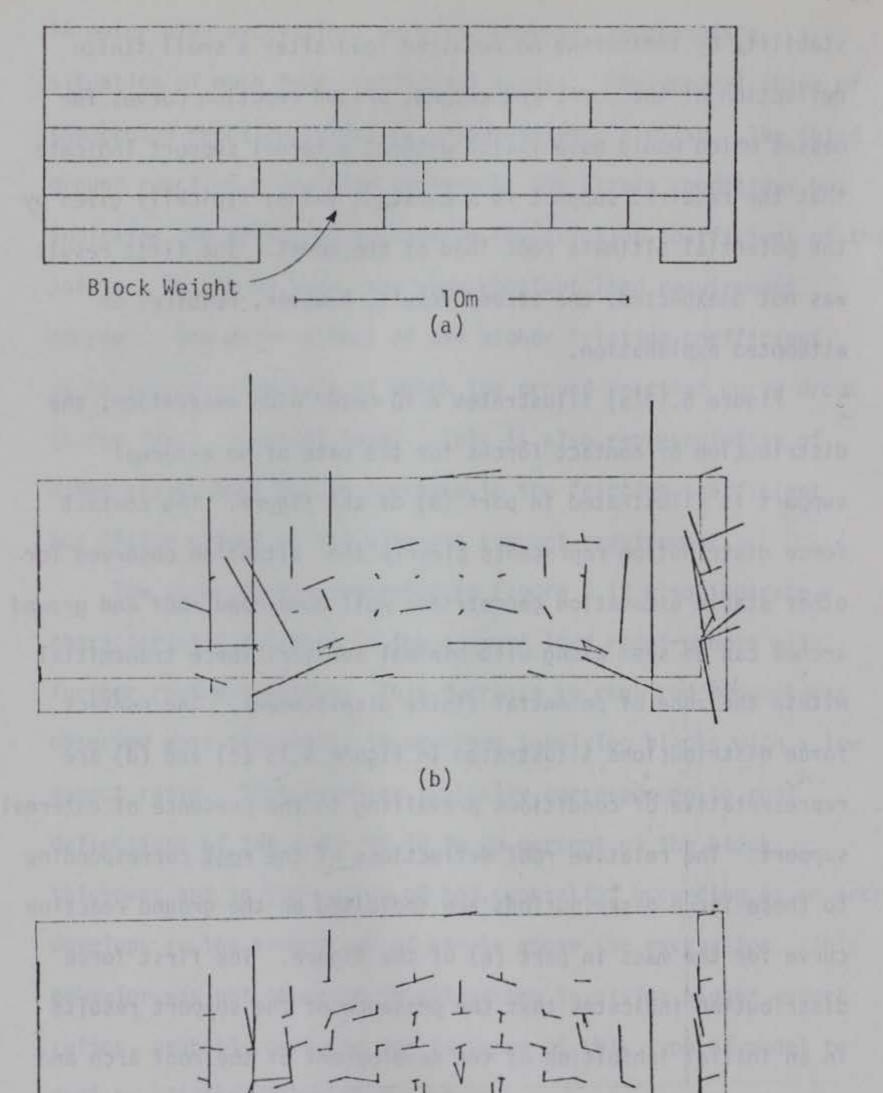
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The presentation of the calculated ground reaction curves has indicated that two general behavior patterns emerged from this investigation: first, ground reaction curves for masses which would have been stable without external support reflect this

stability by indicating no required load after a small finite deflection of the roof; and second, ground reaction curves for masses which would have failed without external support indicate that the required support is a constant value, typically given by the potential ultimate roof load of the model. The first result was not unexpected; the second result, however, requires an attempted explanation.

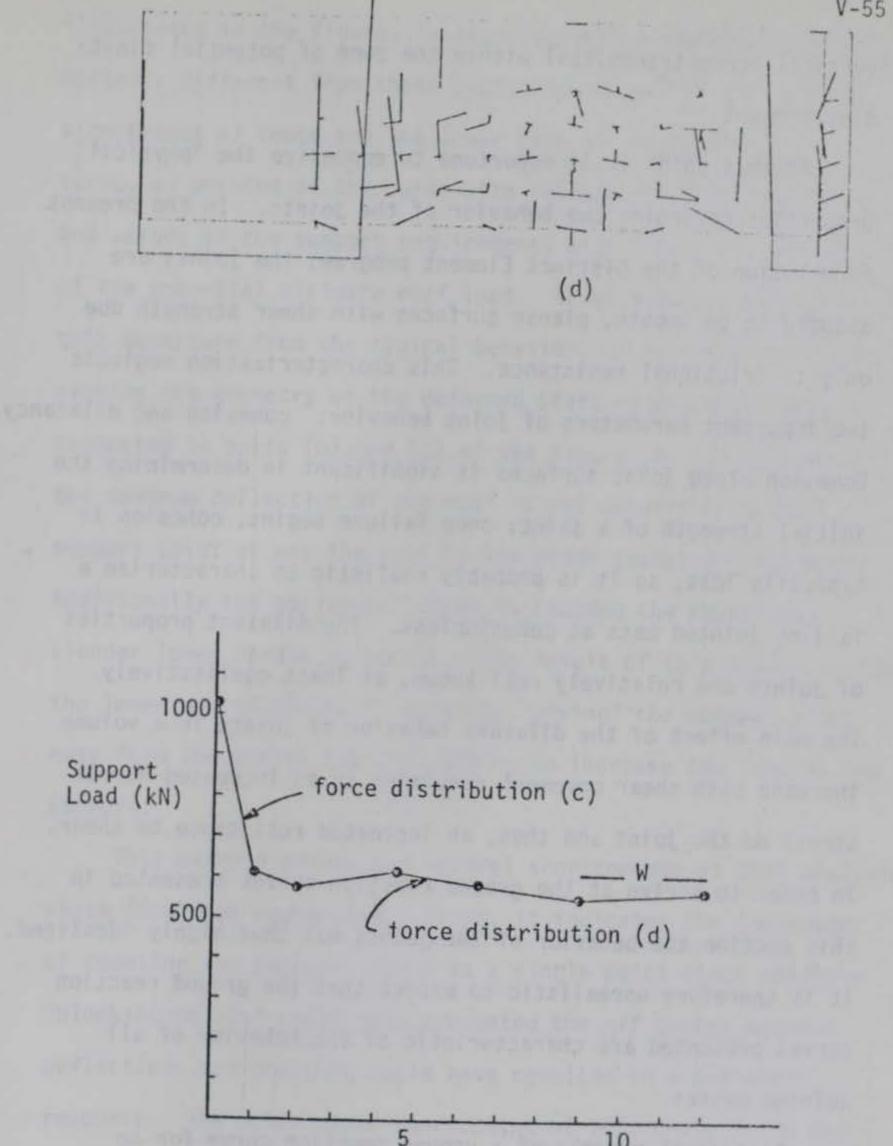
Figure 5.13(a) illustrates a 10 meter wide excavation; the distribution of contact forces for the case of no external support is illustrated in part (b) of the figure. The contact force distribution represents clearly the situation observed for other stable excavation geometries; well developed roof and ground arches can be seen along with minimal vertical force transmittal within the zone of potential finite displacement. The contact force distributions illustrated in Figure 5.13 (c) and (d) are representative of conditions prevailing in the presence of external support. The relative roof deflections of the roof corresponding to these force distributions are indicated on the ground reaction curve for the mass in part (e) of the figure. The first force distribution indicates that the presence of the support results in an initial inhibition of the development of the roof arch and allows vertical force transmittal through the zone of potential

finite displacement. Part (d) of the figure is indicative of conditions on the constant portion of the ground reaction. The roof arch is partially developed, but the presence of the support is preventing the block rotations necessary for minimizing the



(c)

Figure 5.13 Contact Force Distributions for Indicated Model(a); (b) No External Support; (c) and (d) External Support; Relative Deformation Indicated on Ground Reaction Curve (e).



10 Deflection (cm) (e)

Figure 5.13 Continued.

vertical force transmittal within the zone of potential finite displacement.

At this point it is opportune to emphasize the "physical" properties governing the behavior of the joints. In the present formulation of the Distinct Element program, the joints are assumed to be smooth, planar surfaces with shear strength due only to frictional resistance. This characterization neglects two important parameters of joint behavior: cohesion and dilatancy. Cohesion along joint surfaces is significant in determining the initial strength of a joint; once failure begins, cohesion is typically lost, so it is probably realistic to characterize a failing jointed mass as cohesionless. The dilatant properties of joints are relatively well known, at least qualitatively. The main effect of the dilatant behavior of joints is a volume increase with shear movement resulting in an increased normal stress on the joint and thus, an increased resistance to shear. In order to arrive at the ground reaction curves presented in this section the behavior of the joints was thus highly idealized. It is therefore unrealistic to expect that the ground reaction curves presented are characteristic of the behavior of all jointed masses.

As a final example of a ground reaction curve for an

excavation in a jointed rock mass, a situation is presented where the typical, constant ultimate load requirement was not observed. The case under consideration, a 24 meter wide excavation where the jointing defines blocks having an aspect ratio of 0.1, is illustrated in Figure 5.14. The ground reaction curve, also illustrated in the figure, is seen to possess characteristics markedly different from those typically observed. The most significant of these are the lower rate of decrease of the curve, an upswing of the curve with increasing roof defleciton, and values of the support requirements significantly in excess of the potential ultimate roof load. As an aid to understanding this departure from the typical behavior, it is instructive to examine the geometry of the deformed state of the rock mass as indicated in parts (b) and (c) of the figure. As can be seen, the maximum deflection of the roof is not occurring at the support point as was the case in the other geometries examined. Additionally the horizontal force is causing the relatively slender lower strata to buckle. The result of this action is that the lower row of blocks is actually "prying" the support block away form the strata and thus acting to increase the load on the support.

This example points out several shortcomings of this analysis which should be enumerated. First, it indicates the inadequacy of modeling the support system as a single point since multiple "blocking points" could have prevented the off center maximum deflections and possibly could have resulted in a different response. The other major shortcoming of this analysis is the

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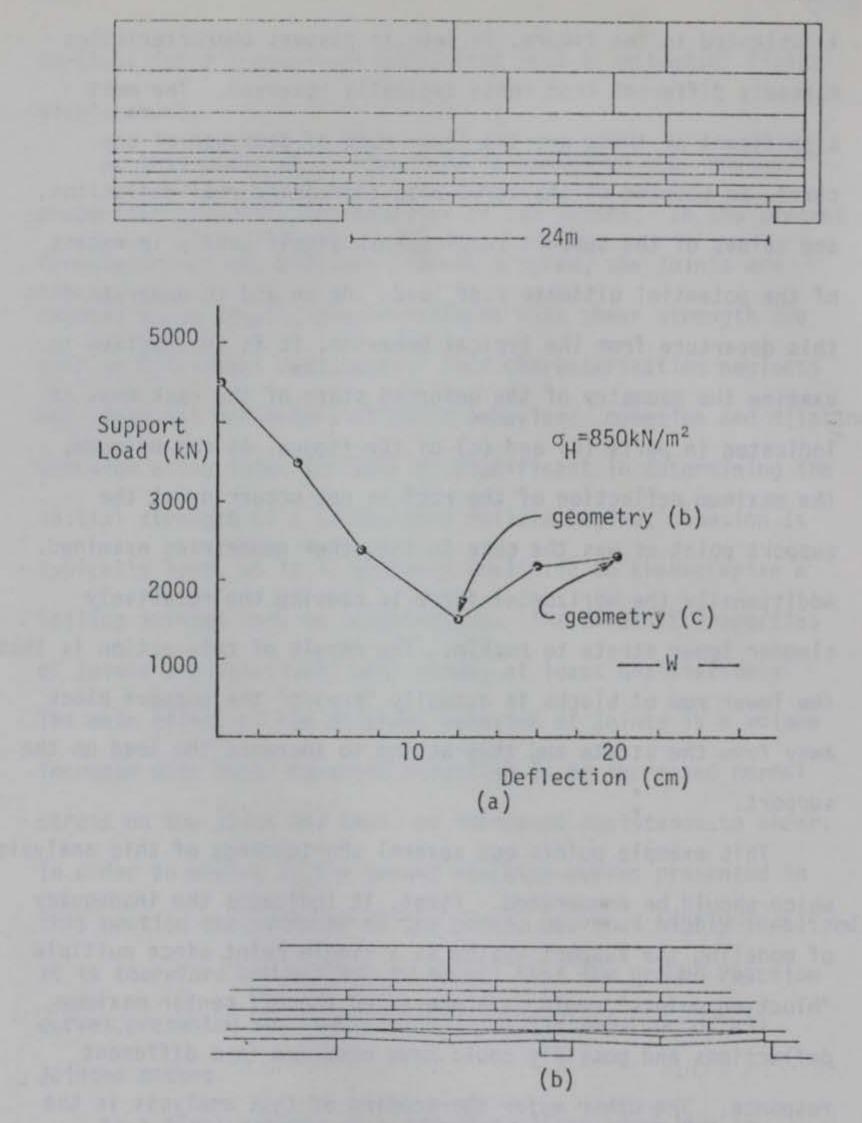
infinite strength of the blocks. In a real situation the

behavior indicated in the figure would probably result in fracture

of the blocks long before the situation indicated in part (c) of

the figure could have developed.





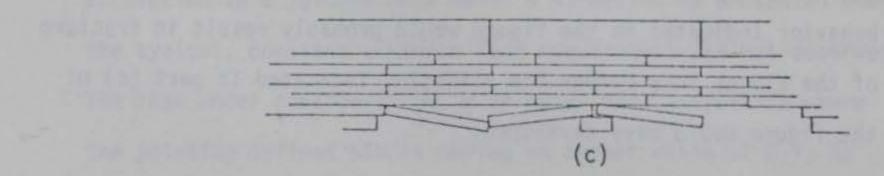


Figure 5.14 Ground Reaction Curve and Displaced Geometries for 24 meter Wide Excavation.

The modeling of jointed excavation roofs presented in this section lead to the conclusion that the ultimate load to be resisted by the support system could be predicted, in the majority of cases, by the potential ultimate roof load described in Chapter 5.2.3. The ultimate loads predicted by the ground reaction curves are summarized in Figure 5.15. Neglecting data from analyses similar to that just described, a relationship between the ultimate support load and the span of the excavation can be seen. This relationship was found to be a function of the aspect ratio of the blocks, but relatively insensitive to the friction coefficient of the joints. The relationship between the support load required and span is given approximately by:

$$L = n B^2$$
 5.14

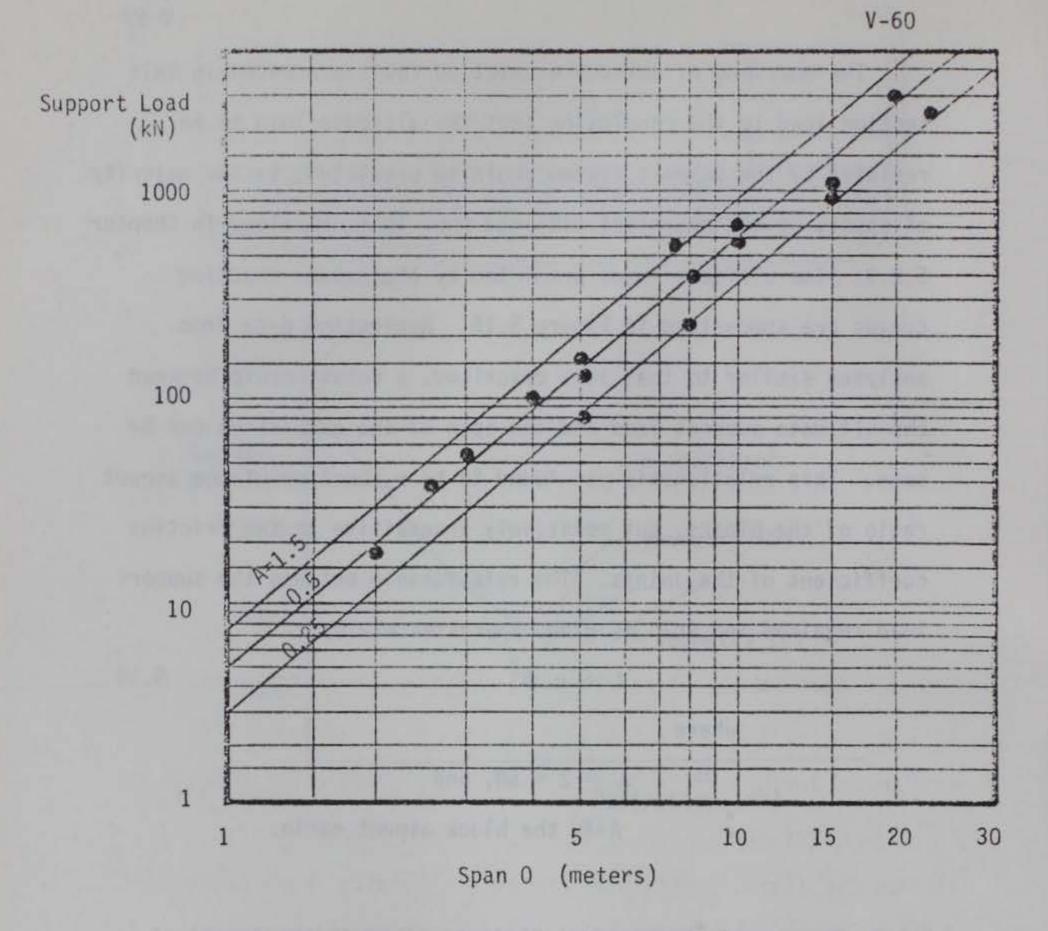
where

n = 2 + 5A, and A is the block aspect ratio.

5.4.2 The use of the Distinct Element method in the design of support systems for excavations in jointed masses

The ground reaction curves presented in the preceeding section indicated that in response to the idealized assumptions of joint behavior utilized in the analyses, the support force required for

stability was seen typically to be a function of the geometric properties of the excavation. In particular, the ultimate resisting force was found to have been given approximately by the potential ultimate roof load, which could be calculated with the aid of



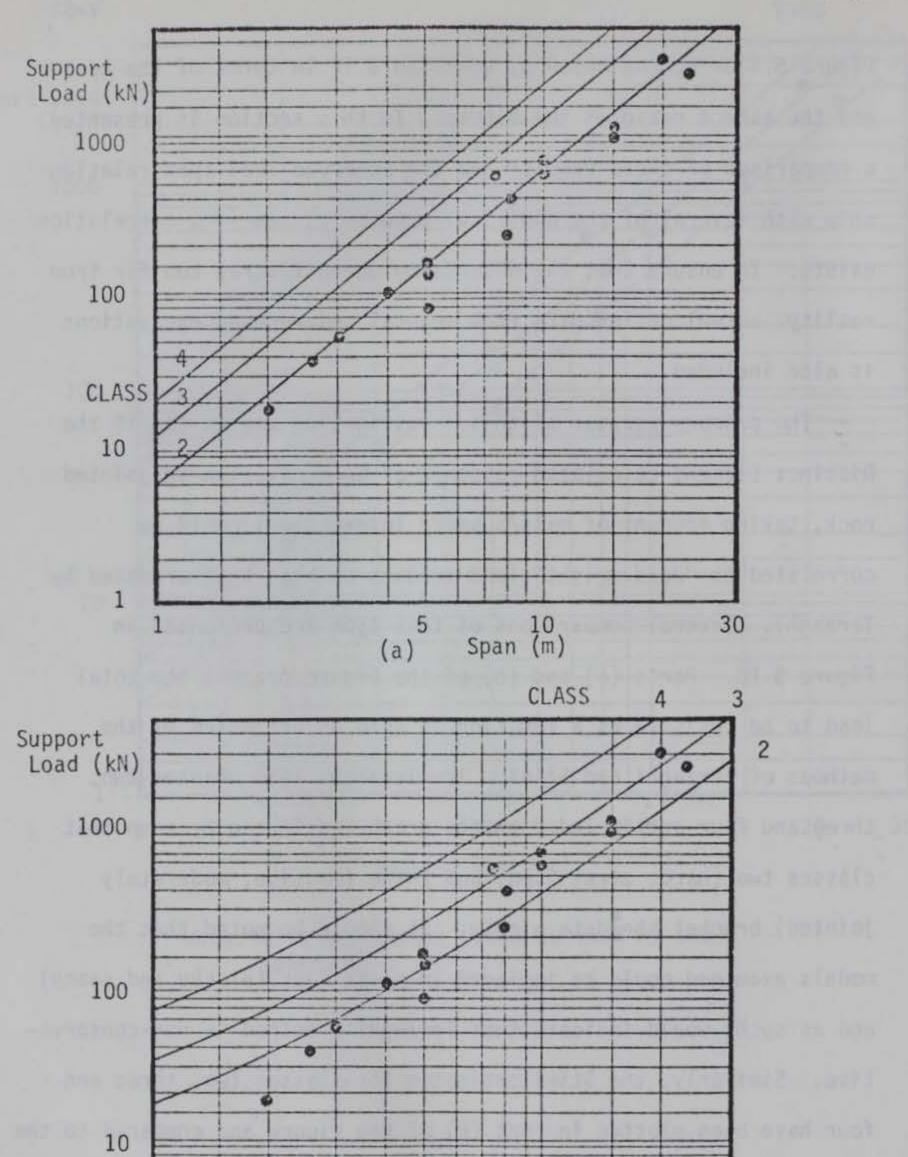
Note: A is the aspect ratio defined by the jointing.

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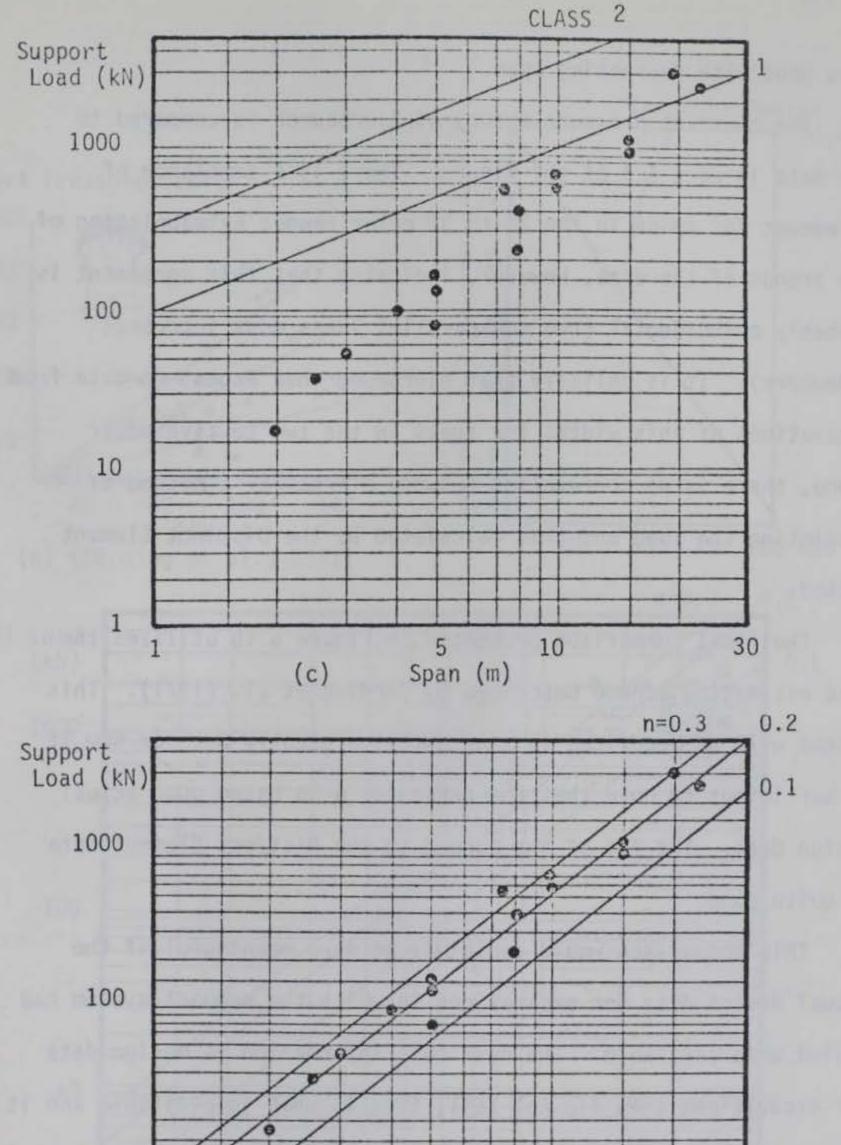
Figure 5.15 Summary of ultimate loads on support system for cases where the mass did not stabilize independently of the support system. Figure 5.4 or approximated by equation 5.17 in terms of the span and the aspect ratio of the blocks. In this section is presented a comparison of these results and the observed load-span relationship with several of the empirical schemes to see if a correlation exists. To ensure that the discussion doesn't stray too far from reality, actual design data from several underground excavations is also included.

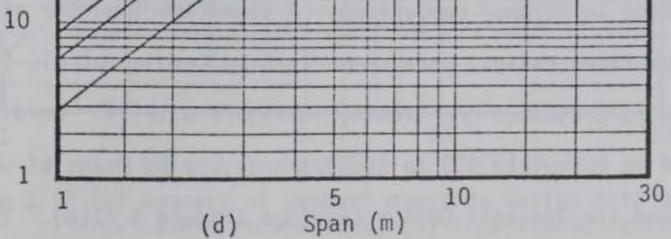
The primary purpose of this investigation was to see if the Distinct Element calculated response of an excavation in jointed rock, taking account of mass/support interaction, could be correlated to "dead weight" load schemes such as that proposed by Terzaghi. Several comparisons of this type are presented in Figure 5.16. Parts (a) and (b) of the figure present the total load to be resisted as a function of span as estimated by the methods of Terzaghi and Stini. The Terzaghi load classes two, three and four are included on the graph and it can be seen that classes two (hard, stratified) and three (massive, moderately jointed) bracket the data nicely. It should be noted that the models examined could be included in class four (blocky and seamy) and as such, would indicate that Terzaghi's method is non-conservative. Similarly, the Stini estimates for classes two, three and four have been plotted in part (b) of the figure and compared to the

Distinct Element responses. Examination of the comparison presented in the figure indicates good agreement with the Stini classes two (nearly stable) and three (lightly broken) for spans greater than about eight meters in width, but the agreement becomes



1 1 5 10 30 (b) Span (m) Figure 5.16 Comparison of Distinct Element calculated required support load with: (a) Terzaghi estimates, (b) Stini estimates.





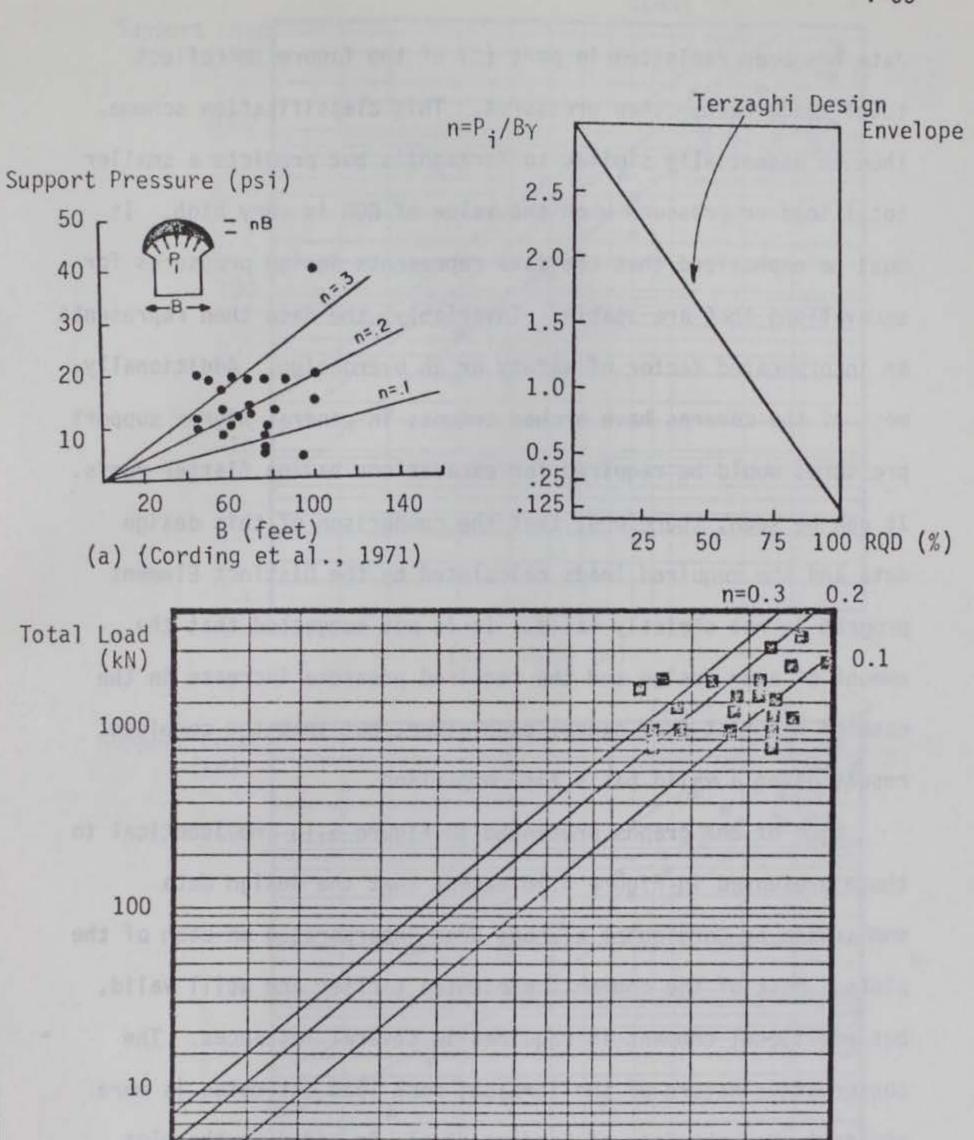
continued, (c) Bierbaumer estimates, (d) Cording estimates. Figure 5.16

less good with decreasing span.

The constant pressure theory of Bierbaumer is compared to the data in part (c) of the figure. There is a semblance of agreement for spans in the 25 to 30 meter range; extrapolation of the trends of the data, however, indicates that this agreement is probably coincidental (two non-parallel lines must intersect somewhere). It is unlikely that Bierbaumer had access to data from excavations of this width; for spans in the two to five meter range, there is no correlation between Bierbaumer's method of predicting the load and that calculated by the Distinct Element method.

The final comparison presented in Figure 5.16 utilizes the load estimation scheme described by Cording et al. (1971). This scheme will be described in some detail presently but for now it is sufficient to note that the parameter n is based upon actual design data. The fit of the curves to the Distinct Element data is quite good.

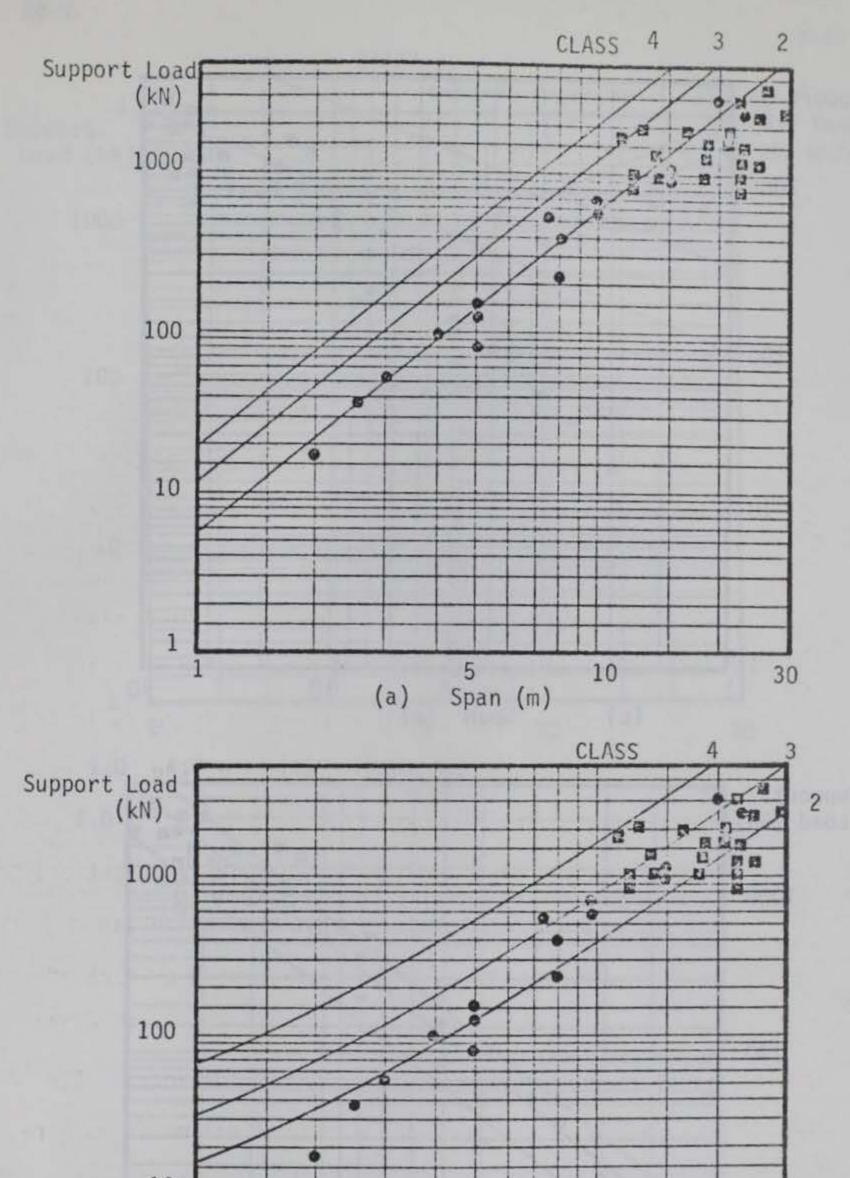
This comparison would certainly be more meaningful if the actual design data for excavations in which the support system had failed were available. The next best information is design data for excavations that did not fail; this is what is available and it will be used in further comparison. A significant number of actual support pressure designs were summarized by Cording et al. (1971); this data is presented graphically in Figure 5.17(a). Cording et al. attempted to correlate RQD to support pressure by means of what they termed the Terzaghi Design Envelope (Figure 5.17(b). This



$1 \frac{1}{1} \frac{5}{(b)} \frac{10}{(b)} \frac{30}{(b)} \frac{30}{(b)} \frac{10}{(b)} \frac{10}{(b)}$

Figure 5.17 (a) Summary of support pressure design data used for cavern excavations, (b) logarithmic representation of total load. data has been replotted in part (c) of the figure to reflect total loads rather than pressures. This classification scheme, then is essentially similar to Terzaghi's but predicts a smaller total load or pressure when the value of RQD is very high. It must be emphasized that the data represents design pressures for excavations that are stable. Invariably, the data then represents an incorporated factor of safety or an overdesign. Additionally, most of the caverns have arched crowns; in general higher support pressures would be required for excavations having flatter roofs. It can be seen, therefore, that the comparison of this design data and the required loads calculated by the Distinct Element program is not strictly valid. It is not suggested that the amount of over design and the required pressure increase in the case of the flat roof cancel each other, but that the combined result gives a valid basis for comparison.

Four of the graphs presented in Figure 5.18 are identical to those presented in Figure 5.16 except that the design data summarized by Cording et al. has been incorporated on each of the plots. Most of the comments presented earlier are still valid, but additional comment is required in several instances. The conservative nature of the Terzaghi rock load estimates is more apparent when the data of Cording et al. is added to the plot. Stini's estimates of the rock load still fit the data quite well for spans greater than 10 meters; unfortunately data for the narrower spans was not available. The rock loads predicted by



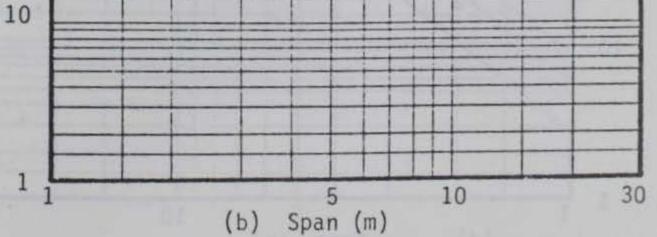
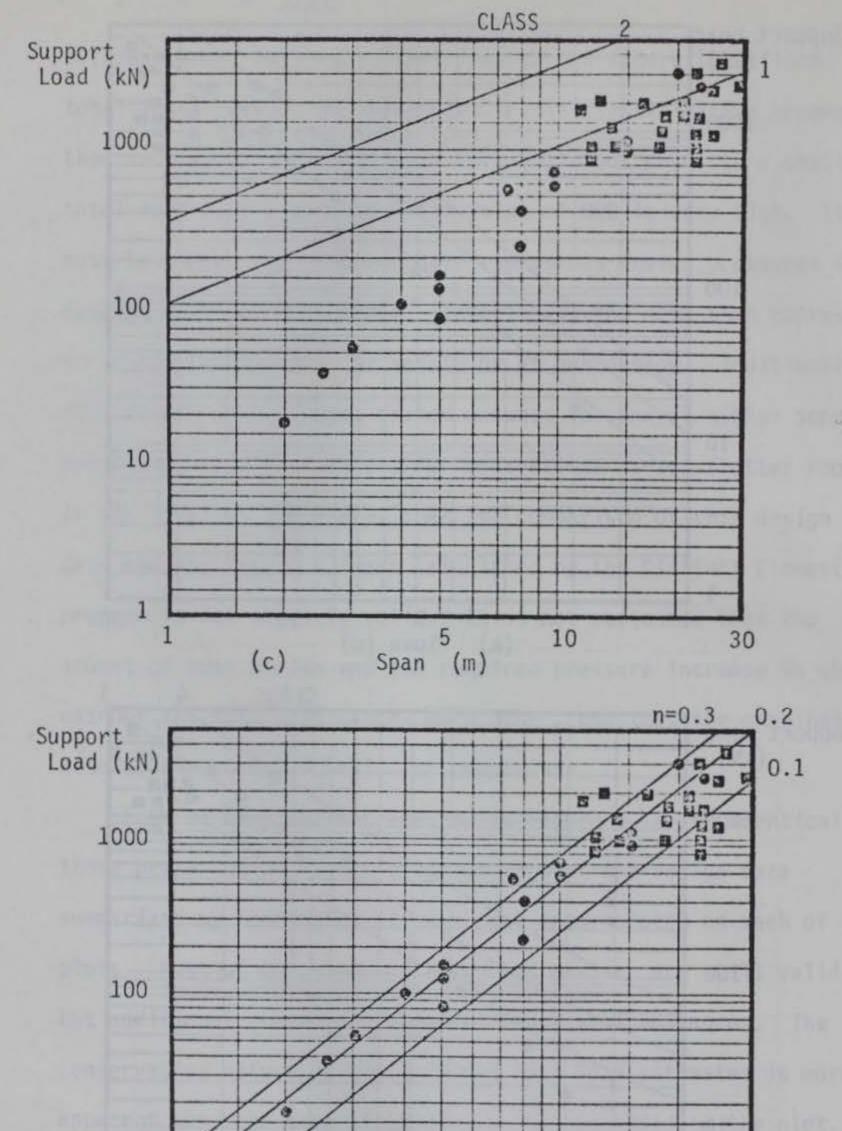


Figure 5.18 Summary of support loads as calculated by the Distinct Element method and reported in the literature Comparisons to metoods of: (a) Terzaghi;(b)Stini;

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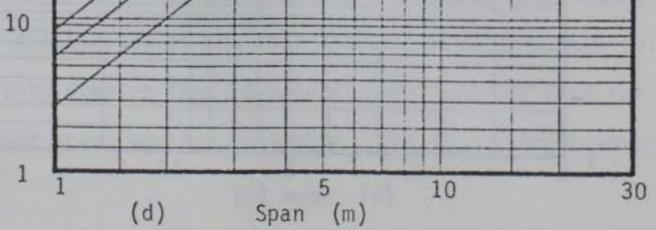
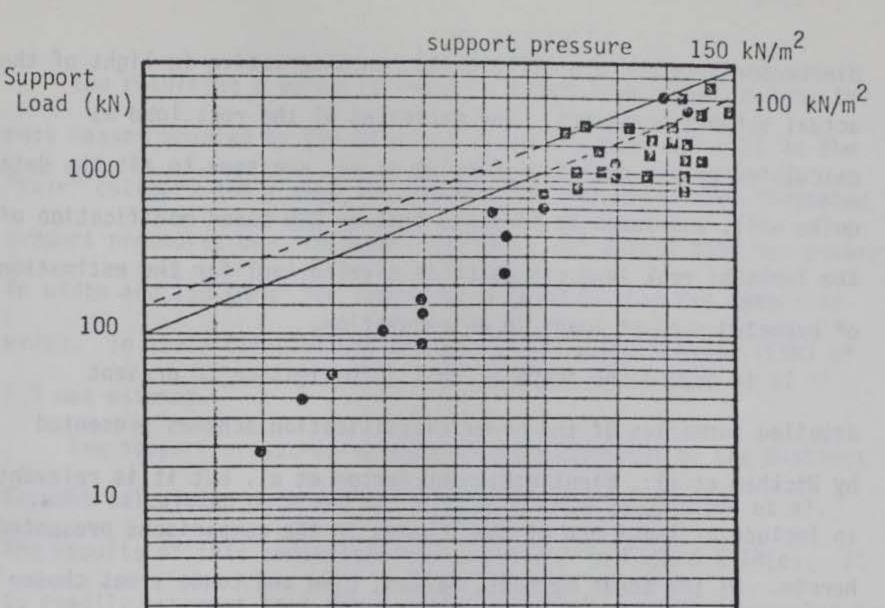


Figure 5.18 (continued) Methods of: Bierbaumer (c); Cording,et al.(d);



1 1 5 10 30 (e) Span (m)

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Figure 5.18 (continued) Method of Barton, et al.

Bierbaumer's method are still quite nonconservative in light of the actual support pressures. The estimates of the rock load as calculated by the method of Cording et al. are seen to fit the data quite well, and seems to indicate that an RQD based modification of the Terzaghi rock load estimates is a valid tool for the estimation of expected support loads in an excavation.

It is beyond the scope of this investigation to present detailed summaries of the newer classification schemes presented by Wickham et al., Bieniawski, and Barton et al. but it is relevant to include at least one of the schemes in the comparisons presented herein. Of the three methods, Barton, Lien and Lunde's was chosen for inclusion for no reason other than that the results are expressed as support pressures. Some familiarity with the method on the part of the reader is assumed.

Barton, Lien and Lunde's classification scheme requires the specification of six input quantities; the values of those quantities thought to represent the Distinct Element modeled geometries are presented in Table 5.5.

Table 5.5 Parameter Values for Rock Mass Quality Q

C)

D)

E)

F)

A)	RQD (Good	to excellent)	75-100%
B)	Joint Set	Number (two joint sets)	4.0

1.0

1.0

1.0

2.5

Joint Set Number (two joint sets) Joint Roughness Number (smooth, planar) Joint Alteration Number (unaltered) Joint Water Reduction Factor (dry) Stress Reduction Factor (low stress) The resulting Q value is found to range from seven to ten; the rock masses modeled by the Distinct Element method all fall in the "fair" category and a need for support is indicated. The indicated support pressures are 100 KN/M² for those spans less than ten meters in width and 150 KN/M² for those spans greater than ten meters in width. In these calculations an excavation support ratio (ESR) of 1.0 was assumed.

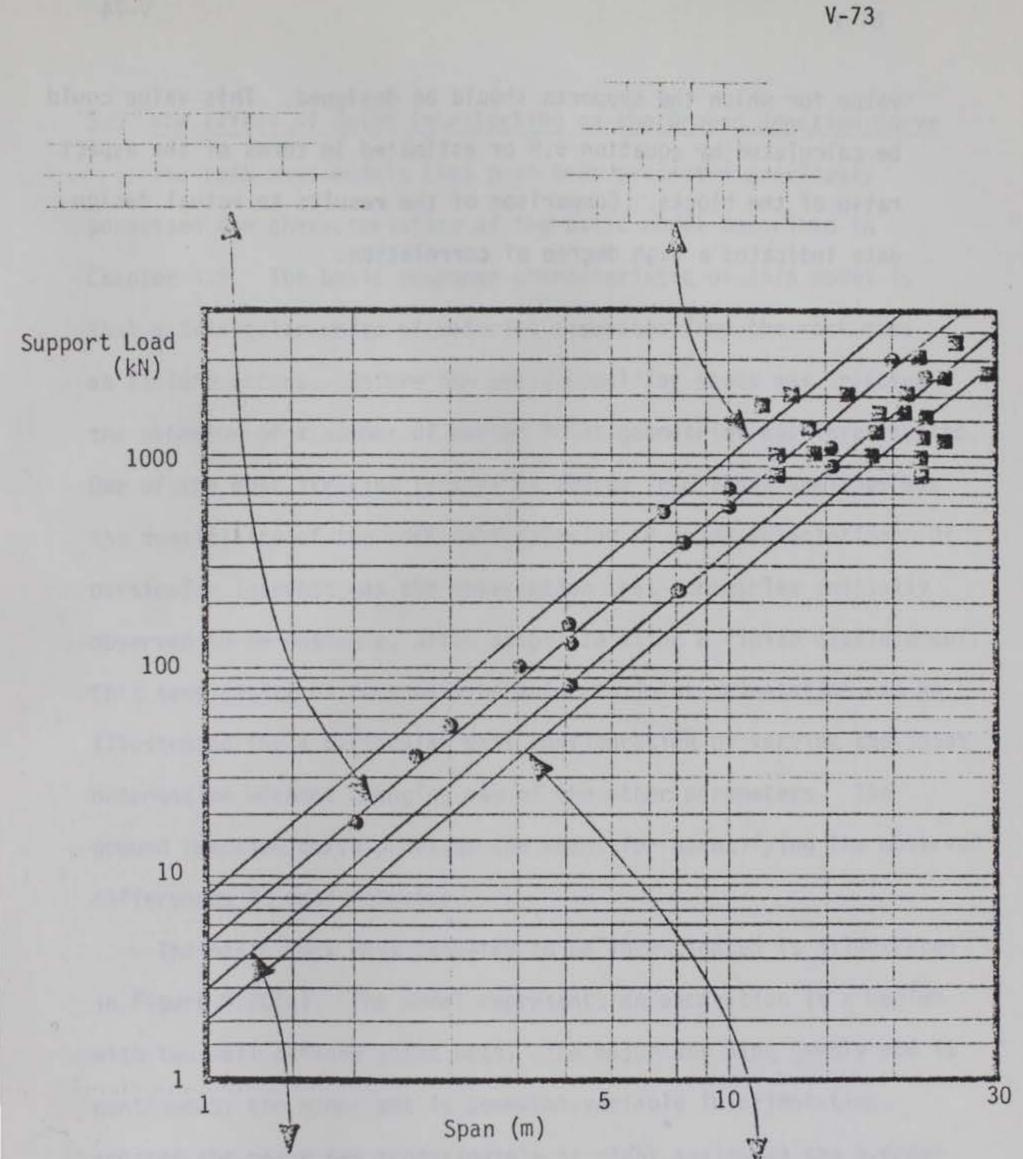
The support pressures calculated were compared to the Distinct Element calculated data and the data presented by Cording et al. The results of this comparison are presented in Figure 5.18(e). It is readily apparent that the constant support pressures suggested by Barton, Lien and Lunde's method do not adequately describe the trends of the data calculated by the Distinct Element method. Furthermore, the support pressures result in total loads that are significantly higher than the data of Cording et al. indicate would be experienced in practice.

The data calculated by the Distinct Element method during this investigation raises one serious objection to the use of the design equation presented by Cording et al. Without exception, all of the geometries modeled using the Distinct Element program had an RQD value of 100 percent. The use of the design equation postulated by Cording et al. would, in this instance, result in a significant underestimate of the amount of required support force. The value of "n" corresponding to an RQD value of 100 percent is 0.1; the majority of the plotted data, both that calculated by the Distinct Element method and that reported by Cording et al. can be seen to lie above the curve corresponding to an n value of 0.1. Perhaps an equivalent RQD based upon seismic velocities could be calculated for the Distinct Element geometries, but it is really outside the scope of this investigation to attempt a correlation of this type.

Figure 5.19 presents a summary of the required support force as a function of span for those masses investigated by the Distinct Element method; also included in the figure is the actual design data summarized by Cording et al. The curves indicating the trend of the data have, in this instance, been calculated using equation 5.14. The presented curves fit the data as well as those suggested by Cording et al.; however, in this case the curves are a function of the aspect ratio of the blocks formed by the jointing. It is not immediately clear that there should be a correlation between RQD and aspect ratio of the blocks. It certainly would be feasible to estimate the block aspect ratio if directionally biased RQD data were available, but RQD data is not typically recorded in this manner.

It was not the intent of this section to deduce a relationship between RQD and the aspect ratio of the jointing; what was desired was computationally based verification of empirical rock load estimation schemes. The properties of the basic model chosen for

investigation indicated that a reasonable estimate of the upper limit to the amount of load to be resisted by the support system could be calculated in terms of the geometric parameters of the rock mass and excavation. The eventual results indicated that this upper limit, the potential ultimate roof load, was actually the



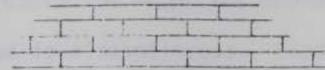
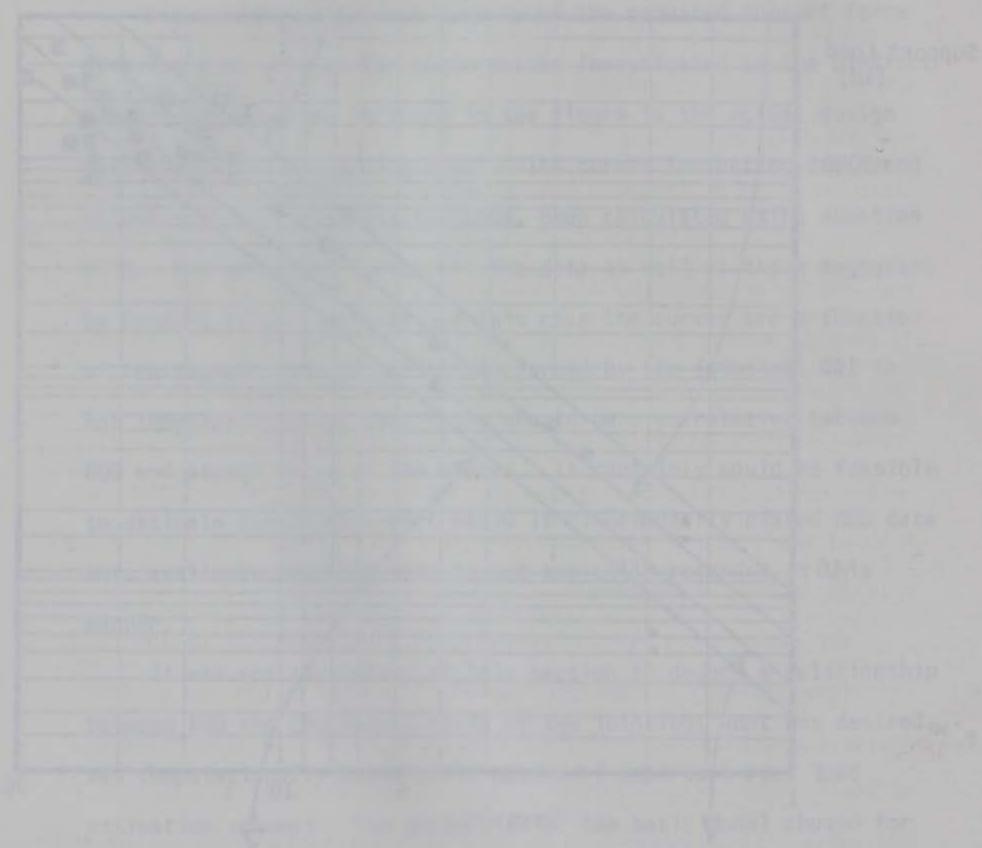
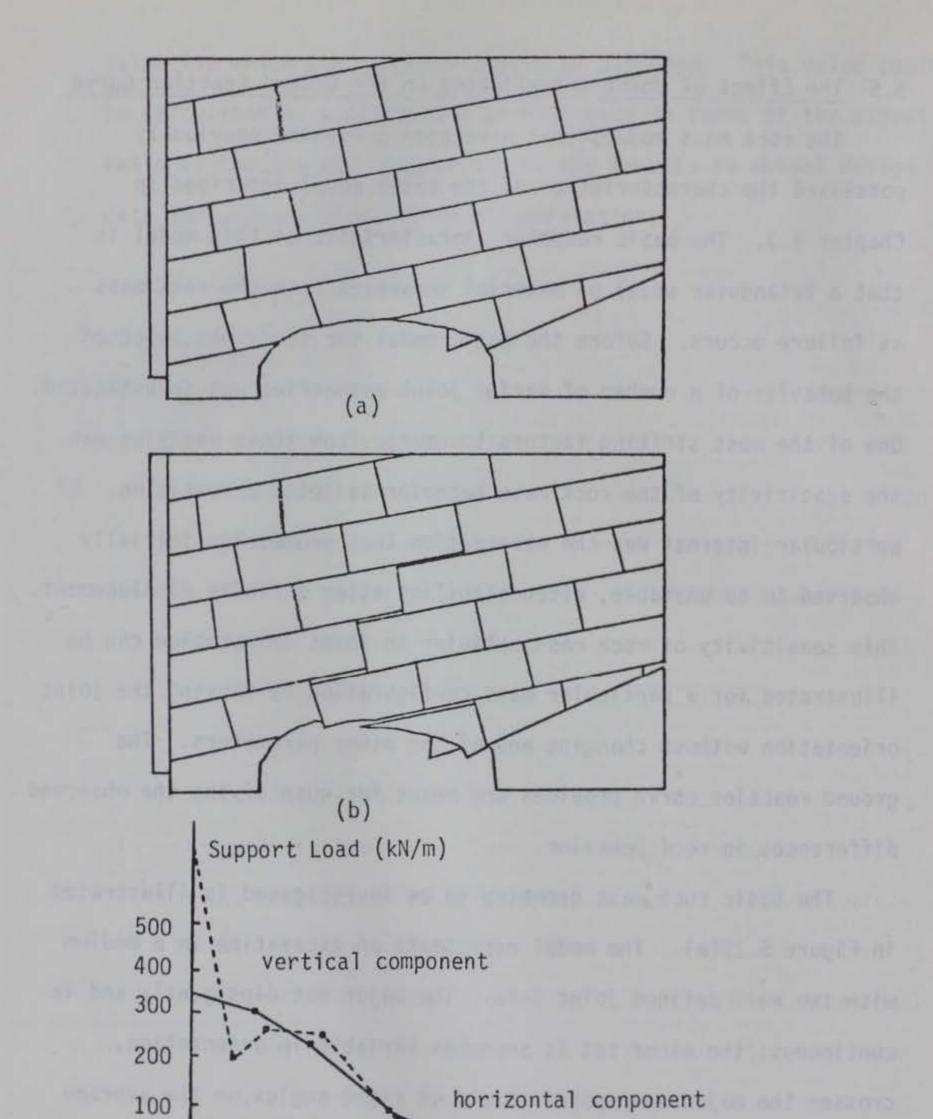


Figure 5.19 Summary of Distinct Element calculated required support loads and design data presented by Cording et al., also illustrated are the various aspect ratios. value for which the supports should be designed. This value could be calculated by equation 5.8 or estimated in terms of the aspect ratio of the blocks. Comparison of the results to actual design data indicated a high degree of correlation.



The rock mass models that have been presented previously possessed the characteristics of the basic model described in Chapter 4.3. The basic response characteristic of this model is that a triangular wedge of material separates from the rock mass as failure occurs. Before the basic model for study was selected the behavior of a number of varied joint geometries was investigated. One of the most striking factors to emerge from those analyses was the sensitivity of the rock mass behavior to joint orientation. Of particular interest was the observation that geometries initially observed to be unstable, often stabilize after a finite displacement. This sensitivity of rock mass behavior to joint orientation can be illustrated for a particular mass configuration by varying the joint orientation without changing any of the other parameters. The ground reaction curve provides the means for quantifying the observed differences in roof behavior.

The basic rock mass geometry to be investigated is illustrated in Figure 5.20(a). The model represents an excavation in a medium with two well defined joint sets. The major set dips gently and is continuous; the minor set is somewhat variable in orientation, crosses the major set approximately at right angles on the average and is discontinuous. Exposed in the upper right hand side of the excavation is an almost triangular wedge of material bounded by joints with a friction angle of 5° ; all other joints have a friction angle of 26.5°. The triangular wedge represents a shear zone and its presence can be expected to govern, or at least severely



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.1 .2 .3 .4 .5 .6 .7 .8 Support Deformation (m) (c)

Figure 5.20

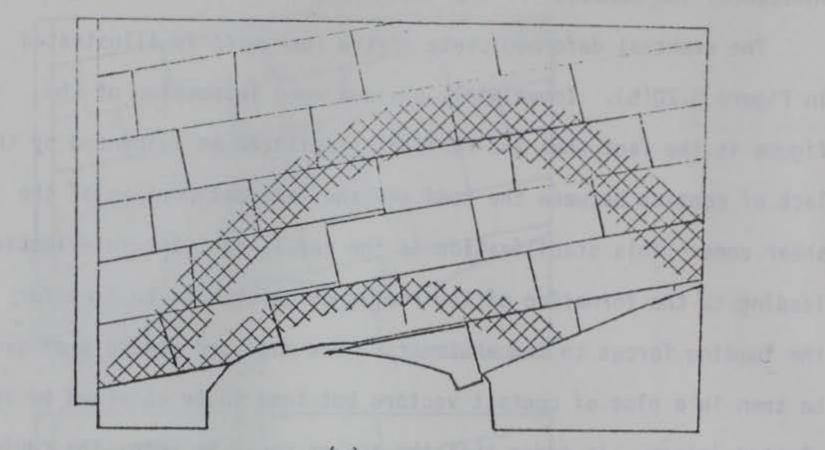
Ground reaction curve for a model where arching acts to stabilize the mass.

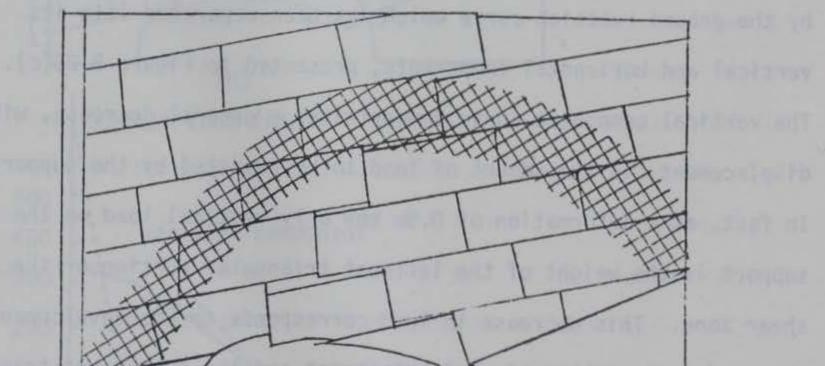
influence, the behavior of the rock mass.

The eventual deformed state of the rock mass is illustrated in Figure 5.20(b). Immediately obvious upon inspection of the figure is the fact that the roof has stabilized as evidenced by the lack of contact between the roof and the leftmost portion of the shear zone. This stabilization is the result of joint interlocking leading to the formation of the roof arch which acts to transfer the loading forces to the abutments. The roof and ground arch can be seen in a plot of contact vectors but tend to be observed by the plotted joints. In order that the arches could be seen, the regions corresponding to the high contact forces have been outlined and shaded; the ground and roof arches corresponding to the rock mass of Figure 5.20 are illustrated in Figure 5.21(a).

A quantitative expression of this arching behavior is indicated by the ground reaction curve which has been separated into its vertical and horizontal components, presented in Figure 5.20(c). The vertical component curve demonstrates a general decrease, with displacement, in the amount of load to be resisted by the supports. In fact, at a deformation of 0.5m the only vertical load on the support is the weight of the leftmost triangular portion of the shear zone. This decrease in load corresponds to the development of the roof arch with vertical displacement and the subsequent transfer of vertical force to the sides of the excavation. The horizontal component indicates that at a deformation of 0.5m the force is practically zero. The reason for this can be seen by reference to the diagram showing the ground and roof arches, Figure 5.21. The







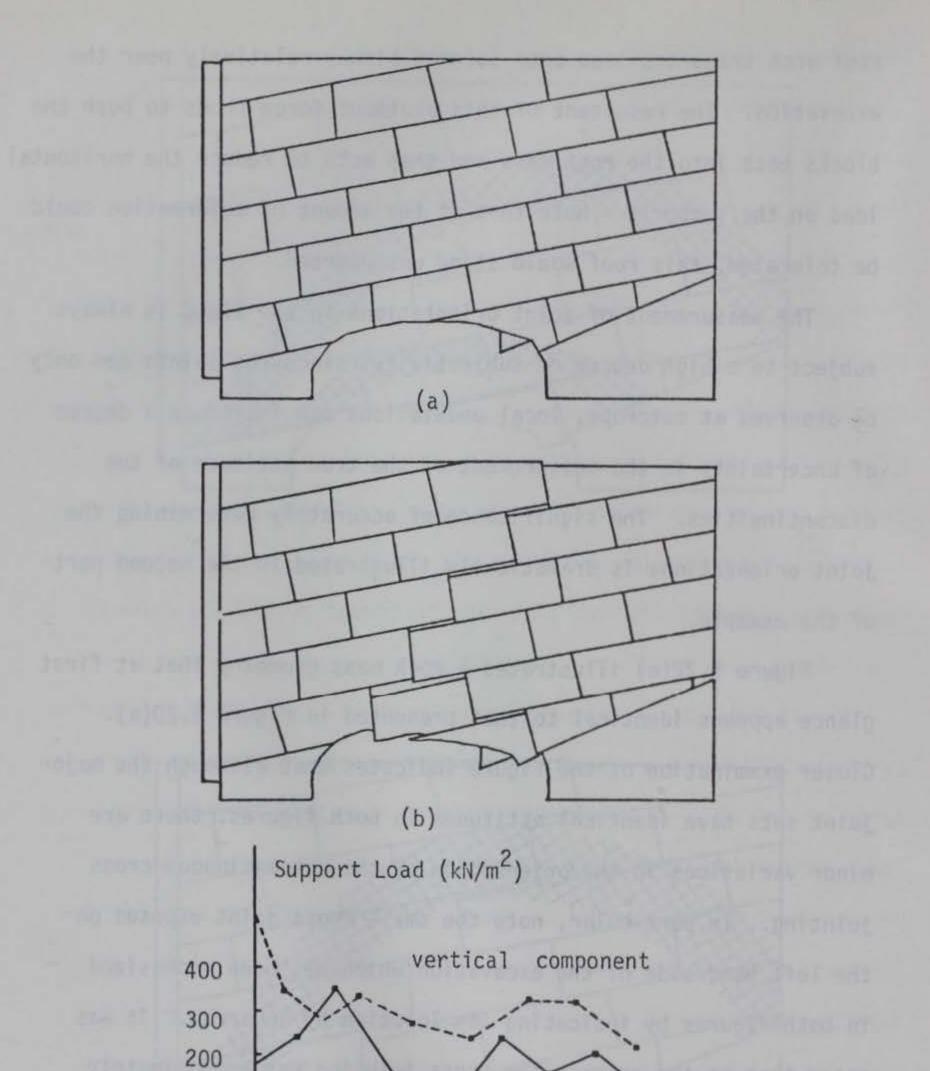
(b)

Figure 5.21 Pressure distributions in: (a) a stabilized roof, (b) a failing roof.

roof arch transfers load onto jointed blocks relatively near the excavation. The resultant of this abutment force tends to push the blocks back into the rock mass and thus acts to reduce the horizontal load on the supports. Note that if the amount of deformation could be tolerated, this roof would stand unsupported.

The measurement of joint orientations in the field is always subject to a high degree of subjectivity; since the joints can only be observed at outcrops, local undulations can introduce a degree of uncertainty in the measurement of the true attitude of the discontinuities. The significance of accurately determining the joint orientations is dramatically illustrated in the second part of the example.

Figure 5.22(a) illustrates a rock mass geometry that at first glance appears identical to that presented in Figure 5.20(a). Closer examination of the figure indicates that although the major joint sets have identical attitudes in both figures, there are minor variations in the orientation of the discontinuous cross jointing. In particular, note the small cross joint exposed on the left hand side of the excavation which has been emphasized in both figures by indicating its loaction by an arrow. It was noted that on the average the cross jointing was approximately perpendicular to the main joint set. An uncertainty of five degrees in the measured orientation of a joint is not a large number, nor are variations in true joint inclination of from five to ten degrees uncommon. Whether the variation between the models arises from errors in measurement or true deviations in joint



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component .2 .3 .4 .7 .8 .9 .5 .6 .1 Support Deformation (m) (c)

horizontal

Figure 5.22 Ground reaction curve for a model where arching does not act to stabilize the mass.

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attitudes is not significant. What is important is the fact that the behavior of the two models changes markedly in response to relatively minor changes in joint orientation.

One stage of the deformation of the model is illustrated in Figure 5.22(b). Examination of this figure indicates a more widespread disruption of the roof than in the previous model but even more importantly, there is continuous contact through the roof down to the support.

Once again the ground reaction curve illustrated in Figure 5.22(c) and separated into its vertical and horizontal components provides the means to quantitatively describe these observations. The most striking dissimilarity in the ground reaction curves is that the second model is characterized by required support loads that do not diminish with increasing displacement. This roof is completely unstable and requires an external support system. The required support is relatively constant with deformation up to a displacement of almost one meter.

The instability of the roof is indicative of the lack of formation of the roof arch. This is indeed the case as can be seen by reference to Figure 5.21(b). The magnitude of the force to be resisted by the supports is limited by the full development of the ground arch. The lack of development of the roof arch prevents the mass from stabilizing and necessitates the emplacement

of an external support system.

It is of interest to compare the actual support loads deter-

mined from the preceeding analyses to the theoretical values as

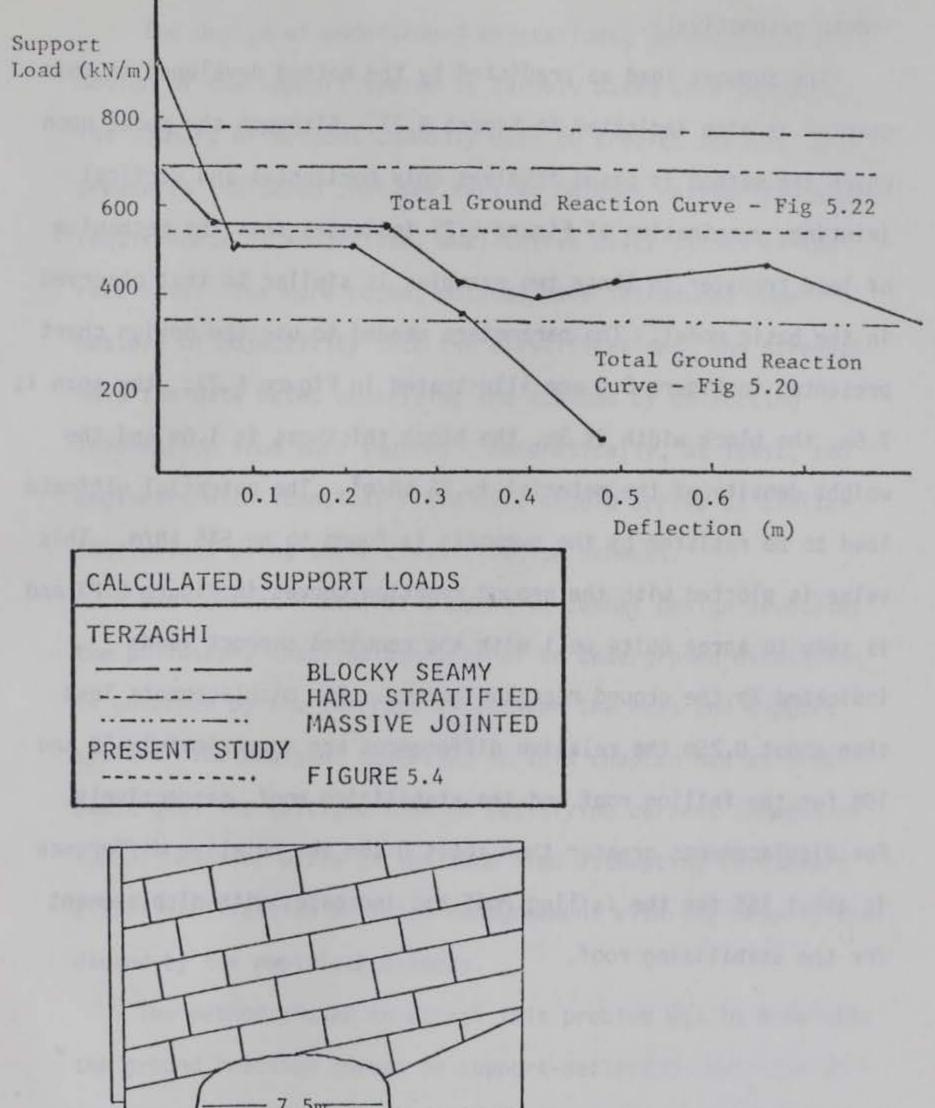
10.4

predicted by Terzaghi's method. The characteristics of the models indicated that the proper classification for these masses was the hard stratified rock category. This category is typified by little resistance against separation along strata boundaries and the weakening of the strata by transverse joints. The moderately jointed rock category requires intimate block interlocking or healed fracture whereas the blocky and seamy category requires blocks which are separated along joints and imperfectly interlocked. The last two categories are actually the limiting cases for the hard stratified rock category.

The sum of the horizontal and vertical components of the ground reaction curves for the two previous examples are plotted in Figure 5.23. Also plotted in the figure are the values of the support load as predicted by Terzaghi's theory.

The constant value of the total support load as calculated for hard stratified rock by Terzaghi's theory is 700 kN/m of tunnel length; compared to the ground reaction curves in Figure 5.23 an over-design is indicated. For displacements less than about 0.25m the relative differences are 25 percent and 30 percent for the failing roof and the stabilizing roof respectively. For displacements greater than 0.25m the relative difference is approximately 50 percent for the failing roof and impressed with displacement for

50 percent for the failing roof and increases with displacement for the stabilizing roof. The relative difference between observed load and predicted load is seen to be significantly greater for the two support load values calculated by the equations for blocky and massive rock masses, which are 800 kN/m and 350 kN/m of tunnel



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Figure 5.23 Comparison of ground reaction curves for a roof that stabilizes after deformation and a roof that fails completely with Terzaghi support loads. length respectively.

The support load as predicted by the method developed in this chapter is also indicated in Figure 5.23. Although the model upon which the method is based involves only horizontal and vertical jointing, examination of Figure 5.21 indicates that the mechanism of load transfer in these two examples is similar to that observed in the basic model. The parameters needed to use the design chart presented in Figure 5.4 are illustrated in Figure 5.23; the span is 7.5m, the block width is 3m, the block thickness is 1.6m and the weight density of the material is 26 kN/m³. The potential ultimate load to be resisted by the supports is found to be 545 kN/m. This value is plotted with the ground reaction curves in Figure 5.23 and is seen to agree quite well with the required support loads indicated by the ground reaction curves. For displacements less than about 0.25m the relative differences are approximately 5% and 10% for the failing roof and the stabilizing roof, respectively. For displacements greater than about 0.25m the relative difference is about 15% for the failing roof and increases with displacement for the stabilizing roof.

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5.6 Summary

The design of underground excavations, particularly the design of the support system is largely based upon precedent. The summary of methods commonly used to predict support load pressures indicated that the earlier methods categorized support requirements by subjective, qualitative descriptions of the rock mass. The more recent methods have introduced some measure of objectivity into the classifications, and strengthened the data bases underlying the schemes by collecting information from more sources. Theoretically, at least, two engineers with identical field data should arrive at similar conclusions using these classification schemes.

One current school of thought in tunnel design advocates the philosophy that the behavior of an underground excavation is governed by the interaction between the mass and support system. The analyses described in this chapter had as their basic goal the multiple task of satisfying current thought on tunnel behavior while at the same time attempting to exhibit either verification or total nonagreement with the results predicted by the empirical methods.

The method chosen to attack this problem was to determine the ground reaction curves or support-deflection behavior of

numerous jointed mass/excavation configurations. In this manner it was hoped to demonstrate that the Distinct Element model solutions would always predict support pressures that were significantly lower than those calculated by the empirical methods, since the predictions of these methods are based upon supporting the total dead weight of a specified volume of rock. For the basic geometry selected for the study, the weight of the material for which it is kinematically possible, neglecting any supporting effects, to move into the excavation, and thus load the supports is easily calculated. It was expected that this potential ultimate roof load would provide a rarely attained upper limit to the necessary value of support resistance indicated by the analyses.

Both of these assumptions were found to be incorrect; in fact, the data indicate that the value for which the supports should be designed is given by the potential ultimate roof load. While this value is typically noticeably smaller than the support loads predicted by the empirical design schemes, there is not enough of a difference to conclude that it has been demonstrated that the use of the empirical methods results in an overdesign.

To understand the reason for the similarity of results, the characterization of the joints must be examined. The joints used at the present time in the Distinct Element method are smooth planar structures which have strength only through frictional resistance. The joints do not possess cohesion. Cohesive resistance is more significant in the initial strength

of a rock mass than in determining the failing behavior. Not much is lost in the analyses of failing rock masses if no cohesion is assumed. The joints also are not characterized by dilatancy. The dilatancy properties of real joints contribute additional strength through volume increase

as shearing occurs. Neglecting the dilatancy of the joints must result in a conservative estimate of the strength. Additionally, in real excavations there is another dilatancy caused by the volume of rock surrounding an excavation moving radially inward. This mass dilatancy also acts to increase the normal force acting on the joints and thus increase the mass strength. The Distinct Element modeled geometries were designed so that only roof deflections were possible and thus neglected this mass dilatancy.

Another limitation imposed upon the analyses described in this chapter is concerned with the joint stiffness. In order that the program could be implemented on a mini-computer, many simplifications needed to be made; one of these was the use of "integer" arithmetic with the burden of watching the signs and decimal points placed upon the programmer (Cundall, 1974). One significant consequence of this was that the joint stiffness turned out to be a function of the problem size. The range of joint stiffness that could be investigated was thus limited. The approximation of the horizontal stress field as a constant load would negate the effects of varying the joint stiffness in any case.

It must be emphasized that the approximations just described are not a consequence of the Distinct Element formulation, but of the mini-computer configuration of the program. These approximations would not need to be made if the program ran in an environment of larger memory on a computer possessing a floating point processor. The implication of the results presented in this chapter can thus be interpreted in one of two ways. By neglecting dilatancy, a correlation was found between the required support force and the potential ultimate roof load. This support force was also found to correlate fairly well with the empirical methods particularly those of Stini and Cording et al. If it can be inferred that the failure to incorporate the dilatancy properties of real joints in the analysis leads to a value of the mass strength that is too low, then it can be concluded that the potential ultimate roof load and thus the empirical methods represent a conservative value of design load.

The second interpretation also follows from the properties of the joints. It is reasonable to expect that the dilatancy properties of joints would play a minor role in situations of relatively low stress. It can thus be concluded that dimensioning the supports to resist the potential ultimate roof load, or using one of the empirical schemes should give the best results in problems involving low stresses.

CHAPTER VI

SUMMARY, CONCLUSIONS AND SUGGESTIONS FOR FURTHER DEVELOPMENT

Before summarizing the results of this investigation, it is imperative that a few sentences be devoted to defining the "ground rules", so to speak, which must govern the discussion which follows immediately. The limitations placed upon joint behavior cannot be overemphasized. The joints within the models utilized in this study were smooth and planar; any shear resisting strength of the joint was due solely to frictional resistance developing as sliding occurred. The joints did not possess cohesive strength; as the cohesive properties are more important in determining the initial strength of the mass, it was felt that little was lost by modeling failing, jointed masses by surfaces having no cohesive strength. The same cannot be said for the fact that the joints utilized did not possess dilatancy characteristics. It is possible that the inclusion of joint dilatancy could significantly affect the resultant mass strength and thus the outcome of many of the analyses reported in this dissertation.

A complete summary of the results of each section is presented at the end of that section; the summary of results presented here will thus be relatively brief. One of the main goals of this dissertation was to demonstrate that the behavior of jointed rock as predicted by the Distinct Element method was realistic. The approach taken to demonstrate the 10.00

validity of the Distinct Element method was based upon comparison to solutions commonly used to describe the behavior of jointed rock masses. The majority of the solution methods chosen for comparison were based upon Limit Equilibrium principles; a basis for selection for comparison was a subjective criterion of how well the solution described the behavior of the model. Thus those solutions selected for comparison are typically simple and the resultant behavior can be intuitively predicted. In all of the comparisons presented in Chapter 3 as well as others presented throughout the remainder of the dissertation, the Distinct Element calculated behavior was seen to correlate quite well with the theoretical solutions.

The second portion of the dissertation described the results of numerous analyses of the behavior of jointed masses by use of the Distinct Element method. The goals of these analyses were to determine those parameters to which the stability of an excavation in jointed rock was most sensitive and to investigate the effects of support interaction in jointed media in an attempt to determine if a rational basis existed for the continued use of empirical design schemes.

The subjects of Chapter 4 were an investigation of the force distributions surrounding excavations in jointed rock masses and

an examination of the stability of unsupported excavations. The topics were approached through numerous models in which the input parameters were varied and the resultant behavior of the model observed. The behavior of the models was illustrated by means of contact force distributions and block displacements plotted on the graphics terminal. The behavior of the models was seen to be governed by force transfer due to the development of arches following block rotations. The stability of an excavation was seen to be sensitive to the horizontal force, the joint friction coefficient and the spacing of the vertical joints. A linear arch analysis neglecting crushing of the blocks and lateral stiffness of the abutments was compared to the behavior as observed by use of the Distinct Element method. Good agreement between theory and observation were noted for single layer models. The theory did not account for the presence of additional shear resistance available in multilayer models and thus there was a poor correlation between theory and observed data.

The investigations described in Chapter 5, on the other hand, were concerned with the behavior of excavations which required externally applied support to maintain stability. The investigations were concerned with the interaction between the supports and the jointed mass and formed the basis for a comparison with different empirical support load prediction schemes. The required supporting force as predicted by the Distinct Element method was obtained through the use of ground reaction curves. These Distinct Element calculated support forces were then compared to the support forces predicted by the empirical methods. Incorporated within this comparison was actual support design data for several underground excavations. The methods which best describe the combined Distinct Element calculated data and design data were seen to be the methods of Cording et al. and the method based upon the potential ultimate roof load described in Chapter 5. It should come as no surprise that Cording et al.'s method fits their data; it is significant that Cording et al's method fits the Distinct Element calculated data and that the support load predictions based upon analyses performed using the Distinct Element method fit the field data as well as is seen. As was noted in the summary of Chapter 5, the incorporation of dilatancy behavior in the joints of the Distinct Element model could significantly alter the results of these comparisons.

The results of the analyses of excavations jointed masses suggest that the Distinct Element method deserves consideration for use in the design of underground excavations. There is not meant to be an implication that all of the information needed to specify a support system for an underground excavation can be obtained by an application of the Distinct Element method. It is only suggested that the Distinct Element method be used as one of the many tools used in the design of an underground excavation.

It is tempting to conclude that a viable design technique would be to analyze a given problem neglecting the dilatant properties

of the joints; using this approach it might be argued that a safety factor would be built into the analysis. However, until the joint dilatancy properties are fully understood it must be recognized that there would be a good deal of uncertainty as to whether or not the safety factor would be one or ten or even one hundred. The data which should routinely be collected during a preliminary site investigation can be utilized in the Distinct Element method to provide preliminary design information. This data would likely include preliminary information on joint spacing, orientation and condition as well as estimates of the horizontal stress state. Using the Distinct Element method, it could quickly be determined if the excavation would be stable or require light or heavy supports. Variations of these input parameters would result in a good idea of how sensitive the excavation stability would be to errors in the assumed values of the input parameters. This analysis could be continuously updated as data from exploratory drilling become available and further refinements could accompany the excavation progress.

This type of design technique is not limited to tunnels; the same data and same procedure are equally applicable to the analysis of slope problems or foundation problems.

These are several reasons that suggest that the method just described is particularly applicable to a class of problems which could be best described as low stress problems. The very nature of the present formulation of the Distinct Element method makes it imperative that it only be applied to problems where the behavior of the mass is controlled by the jointing; this is a characteristic of problems that are near or at the surface. A low stress problem also exists where the frictional resistance of the joints is very low, perhaps due to the presence of clay seams. The investigations

described in Chapter 4 indicated that the material within the zone of potential finite displacement also typically fit the requirements of low stress behavior, although this behavior can be prevented by the presence of high horizontal stresses.

The conclusions to this dissertation must also address the problems encountered due to the mini-computer configuration of the present version of the Distinct Element program. It should be noted from the outset that these are not criticisms of the Distinct Element method itself, but of the equipment upon which the program used in this study presently runs. Foremost of these criticisms must be the time required for a problem solution. The relatively slow computational speed of the mini-computer coupled with the lack of a floating point processor often led to problem solution times which could only be tolerated by someone working toward a Ph.D. Computational times approximately one-twentieth of those encountered during this study could easily be realized on a more powerful computer. However, lost by this implementation would be one of the most powerful capabilities of the Distinct Element program. The insight into the behavior of a jointed mass gained by examining contact force distributions at each time step is often quite revealing. This can realistically only be done on a dedicated computer.

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The amount of computing time required and the limited memory size of the mini-computer also acted to limit the size of the problem that could be investigated. These limitations often resulted in simplified models such as those used to determine the ground

reaction curves presented in Chapter 5. It was noted in Chapter 5 that the idealizations could have masked an important behavior response due to inward movement of the side walls accompanying the roof deflections. This question cannot be resolved until the Distinct Element method is configured on a system possessing a greater amount of memory.

One of the underlying goals of this dissertation was concerned with the utilization of a computer interactive graphics approach to an engineering problem. One particular phase of the project was concerned with developing the graphic interaction capabilities of the present version of the Distinct Element program to the point where an untrained user, particularly one having minimal familiarity with computing techniques, could sit down and use the program to solve simple problems. The solution of this problem was to incorporate a great deal of explanatory material within the program. It is difficult to assess the success of this portion of the project in other than a subjective manner. It did, however, seem as though the majority of those using the program for the first time encountered little difficulty.

Also within the defined goals of this dissertation was the problem of developing a proper perspective as to the applicability of the Distinct Element method. The conclusions drawn are subjective and incorporate material not described in this dissertation. The class of problems most suitable to analyses by the Distinct Element method is characterized by relatively low stress conditions and behavior which is joint controlled. Typical examples of problems meeting these requirements involve slope stability, shallow excavations and foundation behavior. The degree of unconfinement characteristic of these problems ensures that the behavior of these types of problems will be joint controlled. However, the possibility of fracturing of blocks due to local stress concentrations must not be overlooked. It is reasonable therefore to use the analysis obtained by the Distinct Element method in conjunction with an elastic analysis used to determine zones of stress concentration and thus potential fracture. These potential fracture planes can then be incorporated within the Distinct Element method to determine

The dividing line between low stress problems and high stress problems is not clearly defined. It has been noted that the zone of material immediately adjacent to an excavation is under relatively low stress conditions; due to the action of the ground arch the material surrounding the destressed zone experiences much higher stresses. The logical solutions to problems of this type would be either a coupled elastic-Distinct Element program or a modified Distinct Element program which incorporated elastic rather than rigid blocks.

any possible effect.

It is clear from the work typified by Daemen (1975) that highly fractured rock can be modeled by a continuum representation incorporating residual strength properties. It was not possible within the context of the present study, given the limited number of blocks, to determine that point at which the behavior of broken rock ceases to be governed by the directionality imposed by the

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joints and can thus be represented as isotropic. The work described by Bray (1966) does, however, furnish at least a guideline. Bray examined the behavior of jointed masses subjected to an arbitrarily oriented stress field. His results indicated that six independently oriented joint sets were required before the behavior of a jointed mass approximated that of a granular isotropic material. The implication here is that if the material is highly fractured or if the stress conditions are sufficient to fracture the rock it is probably best to adopt a continuum approach.

The research undertaken for this dissertation indicated several areas where further development of the program could be beneficial, and suggested an area of research that could prove to be most rewarding.

The first steps that need to be taken in any further development of the Distinct Element program require faster computational times and a significantly larger computer memory. The results of Chapter 5 were based upon idealized geometries; the typical amount of minicomputer time required to generate one of the ground reaction curves often exceeded two days. This amount of time simply cannot be tolerated if the program is to be accepted as a design tool. The shortcomings of the limited number of blocks were also indicated. The solution to both of these problems is the implementation of the

model on a larger, faster computer.

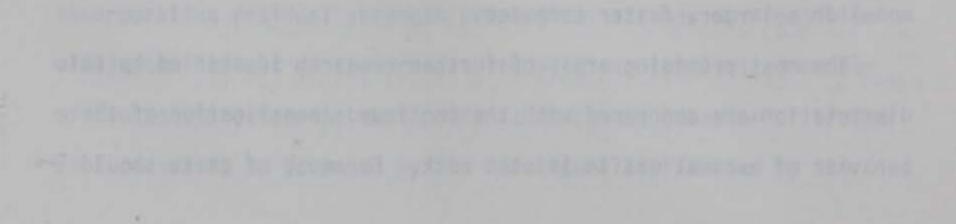
The most promising areas of further research identified by this dissertation are concerned with the continued investigation of the behavior of excavations in jointed rock. Foremost of these should

logically be the incorporation of dilatant behavior of the joints. Additionally, an implementation on a larger computer would allow more blocks per problem and thus a more accurate representation of an underground excavation. This implementation would also allow the incorporation of a stiffness representation of a support system.

This would also lead to a better description of the support system/ mass interaction. It is still felt that, if at all possible, this implementation should take place on a dedicated computer.

The area of research not covered by this investigation which holds promise for a future study is a detailed comparison of the results of observations and careful measurements of physical models and comparable model behavior calculated by the Distinct Element method. This research could form the basis for the incorporation of dilatant behavior in the Distinct Element method as well as providing additional verification of the Distinct Element method through carefully controlled physical testing. In fact, it is easy to visualize a research program that is highly complementary in nature, utilizing a sort of "feedback" system. The Distinct Element method would be useful in the interpretation of the observed data from the physical model while at the same time, the physical model would help to refine the equations used in the Distinct Element

formulation.



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APPENDIX A

THE DISTINCT ELEMENT METHOD

The Distinct Element method is a computer model described by Cundall (1971a) that simulates the behavior of assemblages of rock blocks. The version of the program described by Cundall (1974) forms the basis for the work described in this thesis. Significant features of the program described by Cundall (1974) include arbitrary block shapes, unlimited block displacements and rotations, and a high degree of user interaction. The interaction requires a dedicated computer and centers around a graphic terminal with a cross-hair cursor input capability. The system enables the user to draw a picture of the problem on the terminal and watch the subsequent movement of the blocks as gravity and other loads are applied.

A very thorough presentation of the algorithms implemented in the program, as well as a description of the required hardware, is given by Cundall (1974). The purpose of this appendix is to briefly summarize Cundall's description of the program and note the significant additions to the formulation. Little would be gained by repeating Cundall's descriptions since his report is readily available.

The calculation cycle used in the program is similar to the

one used in most explicit finite difference calculation schemes. Forces arise due to the deformations that occur at corner-to-edge contact points. In each time step of the iteration the incremental shear and normal displacements for a given contact point are calculated using the incremental translational and rotational displacements of the two blocks in contact. The new shear and normal forces acting on the blocks are then calculated from forcedisplacement relationships. All of the contact forces for a given block are then resolved into an equipollent set of forces including a moment acting on the block.

The force and moment sums acting on each block are used to compute translational and rotational accelerations for the block. The accelerations are integrated numerically to obtain block velocities which are then integrated to give the block displacements. With this new set of block displacements the iteration cycle can begin again. Note that if the force and moment sums acting on a block are zero, there will be no acceleration of the block; this is precisely how the program models an equilibrium state.

Before the displacements and accelerations of the blocks can be calculated, however, some method of defining the block geometries must be implemented. The blocks could be treated as "elements" related to defined nodal points as is done in conventional Finite Element analyses. The input would thus consist of numerous cards containing nodal point and element data; anyone who has attempted this to define a mesh for a Finite Element analysis is acutely aware of the frustration that results from trying to "debug" such

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a mesh. The approach adopted by Cundall (1974) and implemented in the program used for the research described in this dissertation overcomes the difficulties associated with mesh generation. The actual rock mass geometry, as defined by the jointing, is drawn on the screen of the CRT. All calculations necessary to determine the significant coordinates are thus performed by the program. The structure of the program is governed by the size limitations imposed by the mini-computer; the actual program consists of three overlays which correspond to the three main calculation phases of the program.

Phase 1 of the program governs the interactive dialog by which the lines defining the block geometry are created. A flow chart for this section of the program is given by Cundall (1974); the flow chart is essentially valid for the present configuration of the program. Care was taken so that the changes to Phase 1, which will be described presently, did not alter the program sequence or execution.

The two main changes made in the Phase 1 section of the program are concerned with the format of the data input and the storage and subsequent retrieval of data files. Whereas the initial version of the program used only the cross-hair cursor of the CRT for input, the present version of the program uses a graphic tablet ("digitizer") and a numeric input scheme as well. The three routines are virtually identical and, in fact, use only one set of coding. Whichever routine is active at a given time is noted by the value of the variable KODE: KODE = -1 signifies that the numeric input routine is selected; KODE = 1 signifies that the graphic tablet is in use; and, KODE = O signifies that the cross-hair cursor is being used for input. All three input methods may be used for a single problem. Potential users wishing to implement the modified version of the program need only supply software for the graphic tablet (Subroutine DIGIT). It should be noted that the numeric input routine contains a scale factor. In this manner, actual field

coordinates may be used as input, and divided so that they meet the program requirements (see Cundall, 1974).

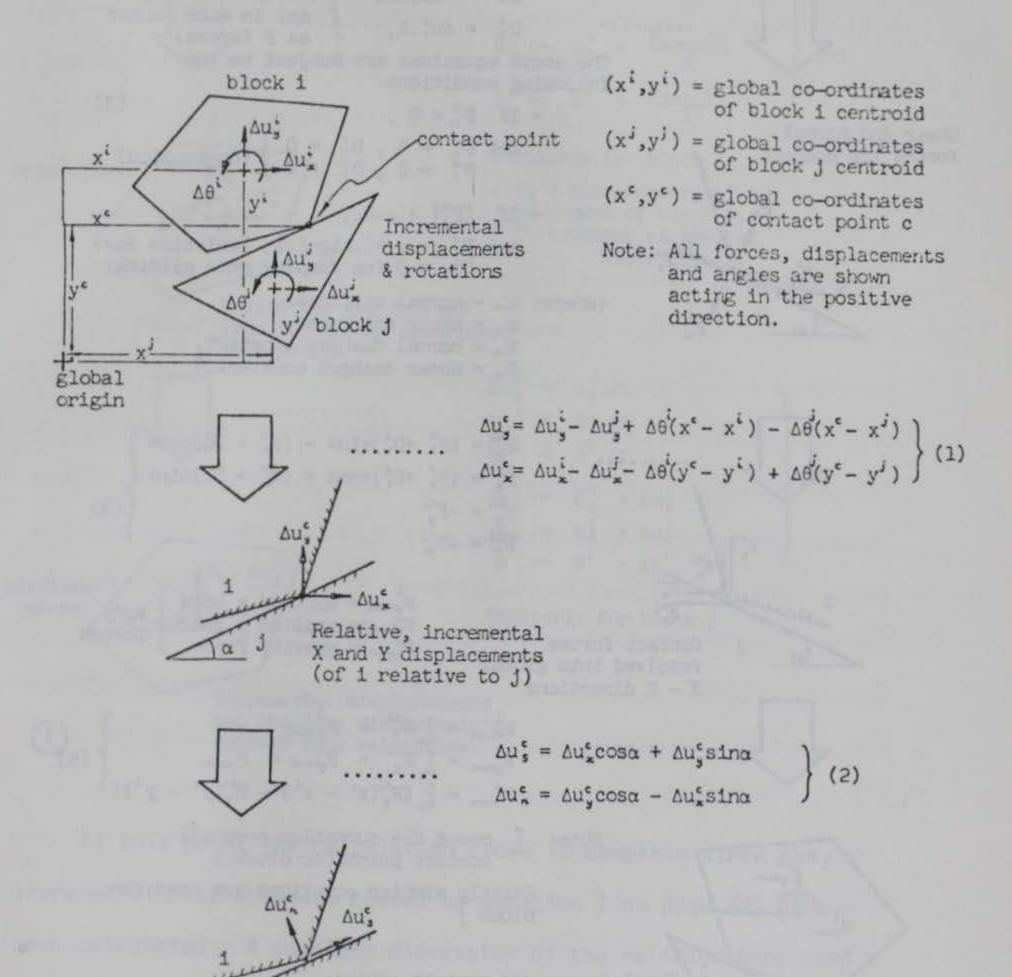
The second major change in the Phase 1 program enables users to store data files consisting of line segments and coordinate data. To do this, the common blocks are written to or read from the Linc tape units. The operation is straight forward; line 57 of the program (see Appendix C) LIST (3) = 13286 is simply a "password" to prevent garbage from being read as a data file.

The second overlay, Phase 2, is unchanged from Cundall's (1974) original listing. This is the routine that scans the line segments created in Phase 1 of the program and converts the line segments to closed areas. A flow chart for this routine is presented by Cundall (1974).

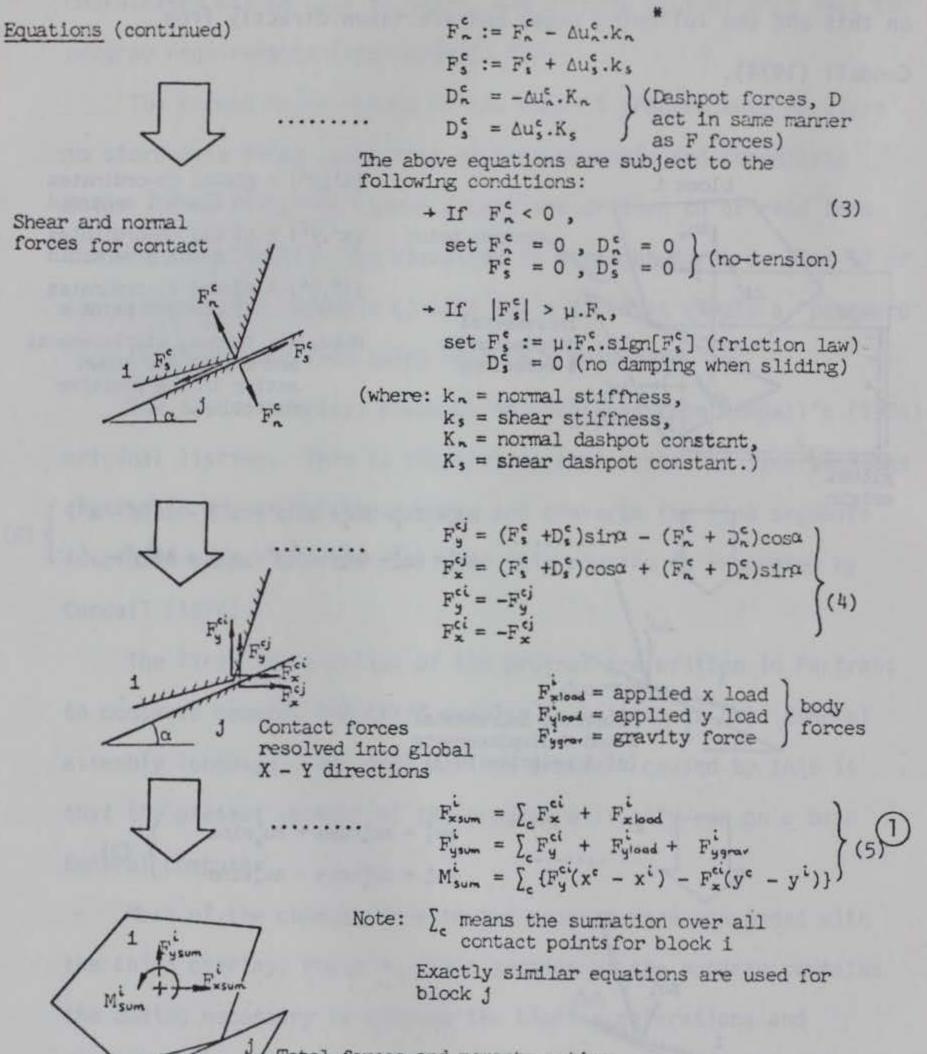
The first two overlays of the program are written in Fortran; to conserve memory, the third overlay is written in Data General assembly language. The only serious drawback caused by this is that the present version of the program will only run on a Data General computer.

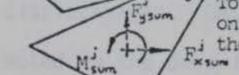
Most of the changes made to the program were concerned with the third overlay, Phase 3. This section of the program contains the coding necessary to compute the block accelerations and

displacements. Detailed descriptions of the modifications will be noted in the descriptive summary of the Phase 3 subroutines to be presented shortly; the main calculation cycle, however, remains essentially unchanged. The equations used in the main calculation cycle are summarized on this and the following pages and are taken directly from Cundall (1974).



Relative, incremental normal and shear displacements (of i relative to j)



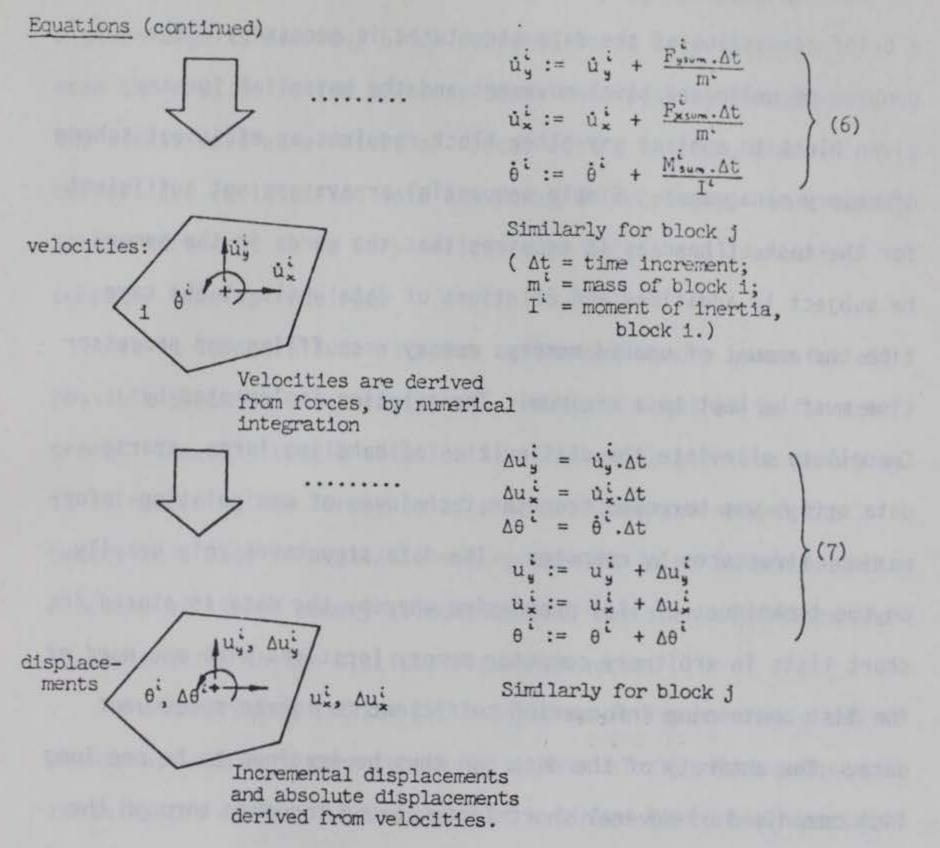


Total forces and moments acting on block i found from the sum of Fi the contributions of each contact.

* The symbol := means "replaced by"

) The formulation of equation 5 differs slightly when joint water pressure is present (see page A-22).





At this point the calculation cycle is complete since the incremental displacements needed by equation 1 on page A-5 have been calculated. A complete discussion of the relationships used

in equations 1 - 7 is given by Cundall (1974). The algorithms used to derive the coordinates and angles used by equations 1 and 2 are also presented.

As a prerequisite to the discussion of the Phase 3 subroutines, a brief discussion of the data structures is necessary. The problem of unlimited block movement and the potential for any given block to contact any other block requires an efficient scheme of memory management. Simple sequential arrays are not sufficient for the task at hand as it requires that the words in the memory be subject to additions and deletions of data while at the same time the amount of unused memory, memory reshuffling and processor time must be kept to a minimum. The solution implemented by Cundall to alleviate the difficulties of handling large, sparse data arrays was borrowed from the techniques of manipulating information structures by computer. The data structures rely heavily on the techniques of list processing whereby the data is stored in short lists in arbitrary computer memory locations with one word of the list containing information sufficient to locate subsequent data. The entirety of the data can thus be imagined to be one long list comprised of several short lists strung together through the memory. The reader who requires exact details concerning the implementation of the list processing techniques is advised to consult Cundall (1974) pages 62 - 72. All that will be presented herein is a brief overview of the list processing implementation

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and a description of the format of the data structures used in the present formulation of the program.

The storage requirements for a given block model due to the

problem of allowing any block to touch any other block are overcome by a list scheme. All block corners are classified into coarse boxes covering the screen area. When the program needs to know if a given edge is near any block corners, it is only necessary to scan the area delimited by those boxes encompassing the edge. As the blocks move as a result of forces acting on them, their corners are reclassified into new boxes if necessary. This boxing scheme turns out to be very efficient as only a small amount of computer time is required.

It is impossible to allocate sufficient memory space for all possible block to block contacts - the space required is far too great. The only viable solution is a method to allocate memory as it is needed by the formation of a new contact and return the memory to a pool of available memory when it is no longer needed. A scheme of linked memory allocation provides such a solution and is implemented in the Distinct Element program.

In the program a fixed group of words is reserved as a set of pointers; each word corresponds to a given block. Each pointer contains the address of the start of a linked list of all contacts for the block associated with that pointer. Another list is used to store all of the memory which became "dead" once a contact was broken. When a new contact is detected by the program the program first checks the list of dead contact space. If space exists it

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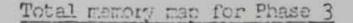
is used, otherwise, previously unused memory at the high end of core is allocated. The following pages describe in detail how the data is organized in the computer memory. The first page following shows a total memory map illustrating the four main parts of the

memory. These are:

- a) the program
 - b) the sets of data pertaining to each block
- c) the pointers and data necessary for the "boxing" scheme, and
- d) the data sets and pointers pertaining to the contact between blocks

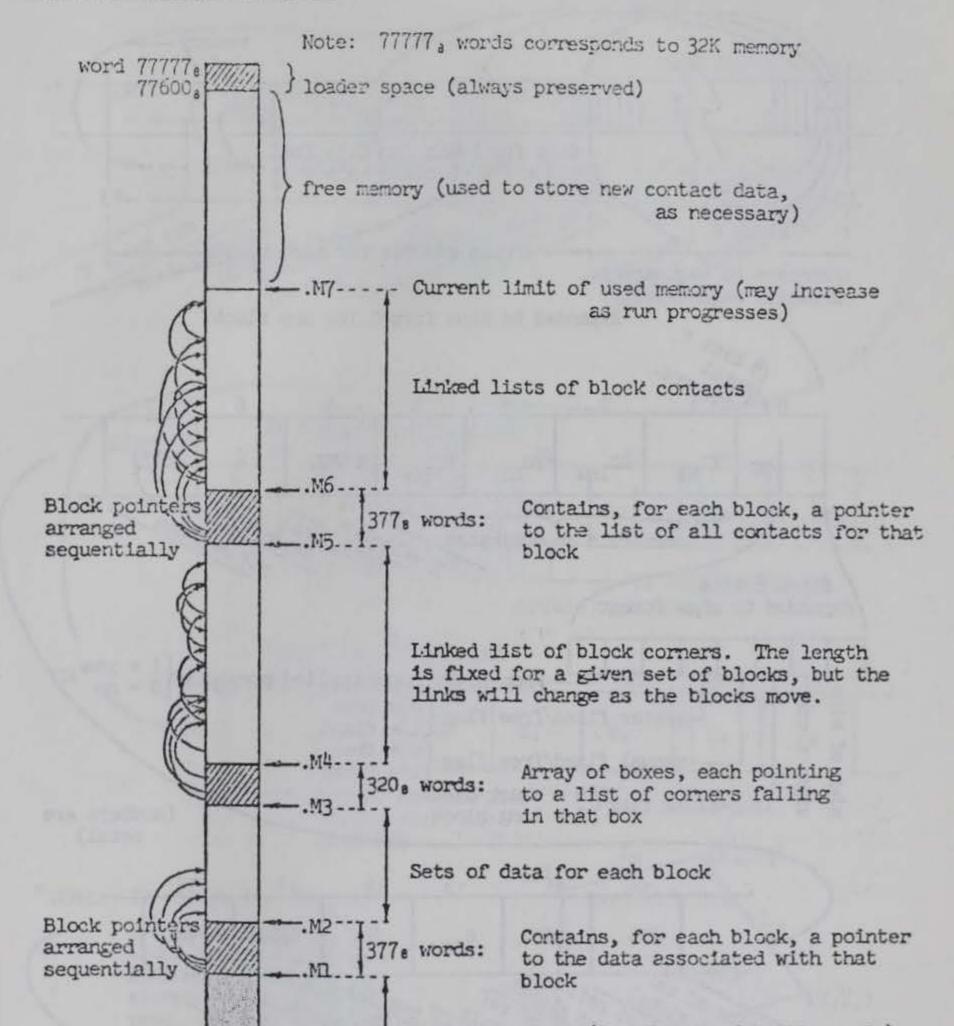
The subsequent pages illustrate expanded forms of groups b, c, and d to show in detail the structure of each list.

The present formulation of the program utilizes another linking scheme to store the data pertinent to applied joint water pressures when they exist. The format of data lists used in this scheme is also illustrated. There are two other linked lists threaded through the memory that must be mentioned; these are the "empty" lists used to reference previously used memory space that is now free for re-use. Memory is made available whenever a block contact is broken or when a pressure segment is deactivated. The two empty lists and the joint pressure lists are referenced by global memory pointers and make use of whatever memory is available. Adding or reclaiming a group of words from the empty lists is simply a matter of reshuffling the link bits and is illustrated by Cundall (1974).



400 8

word 0

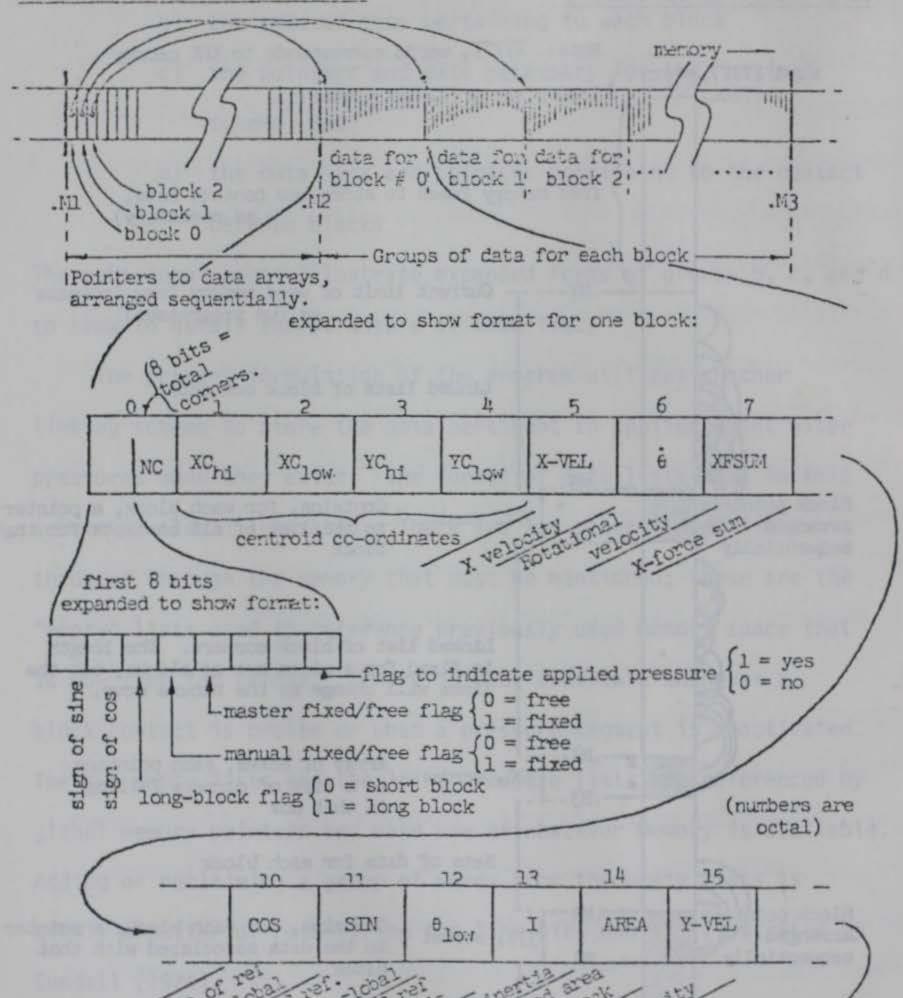


Phase 3 program (length around 15000, words)

Page zero: reserved for storage of global symbols

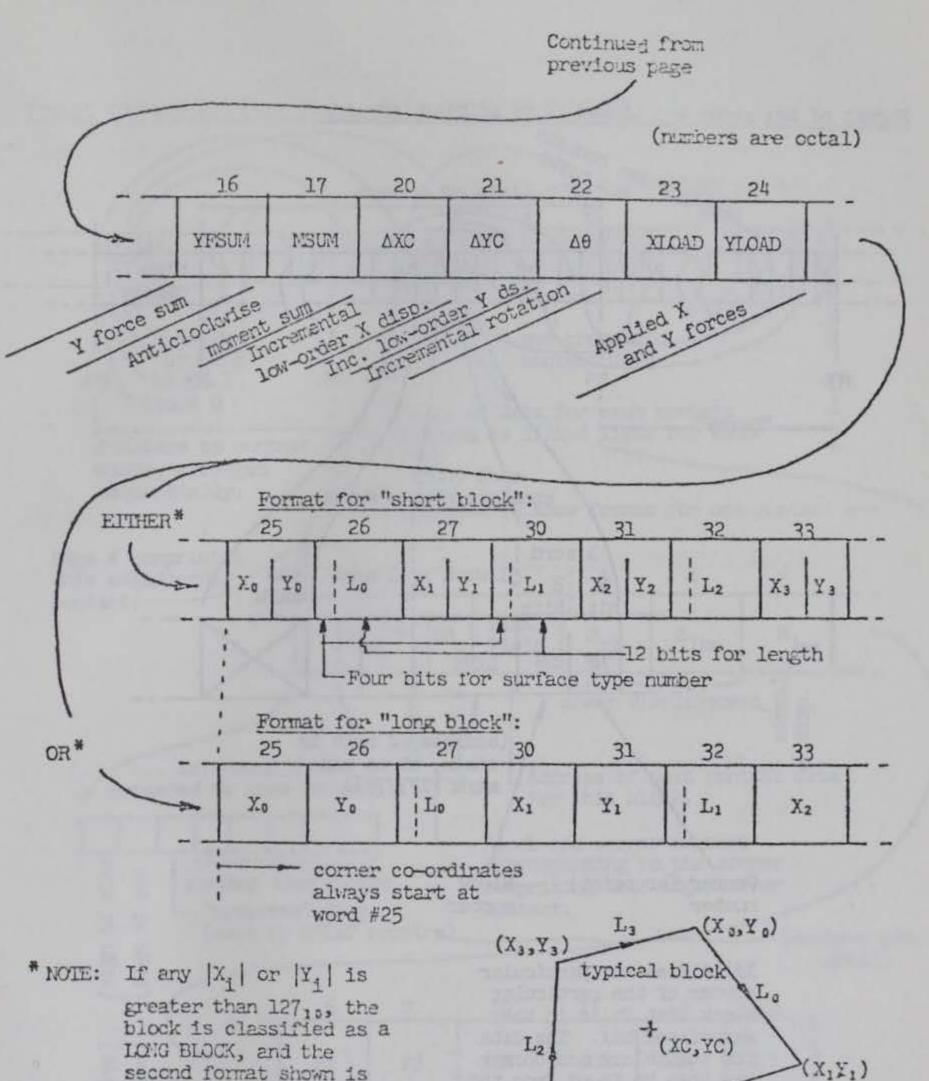
Note:

.Ml, .M2 etc are the global symbols that refer to the pointers to the memory locations shown Format of date arrays for blocks



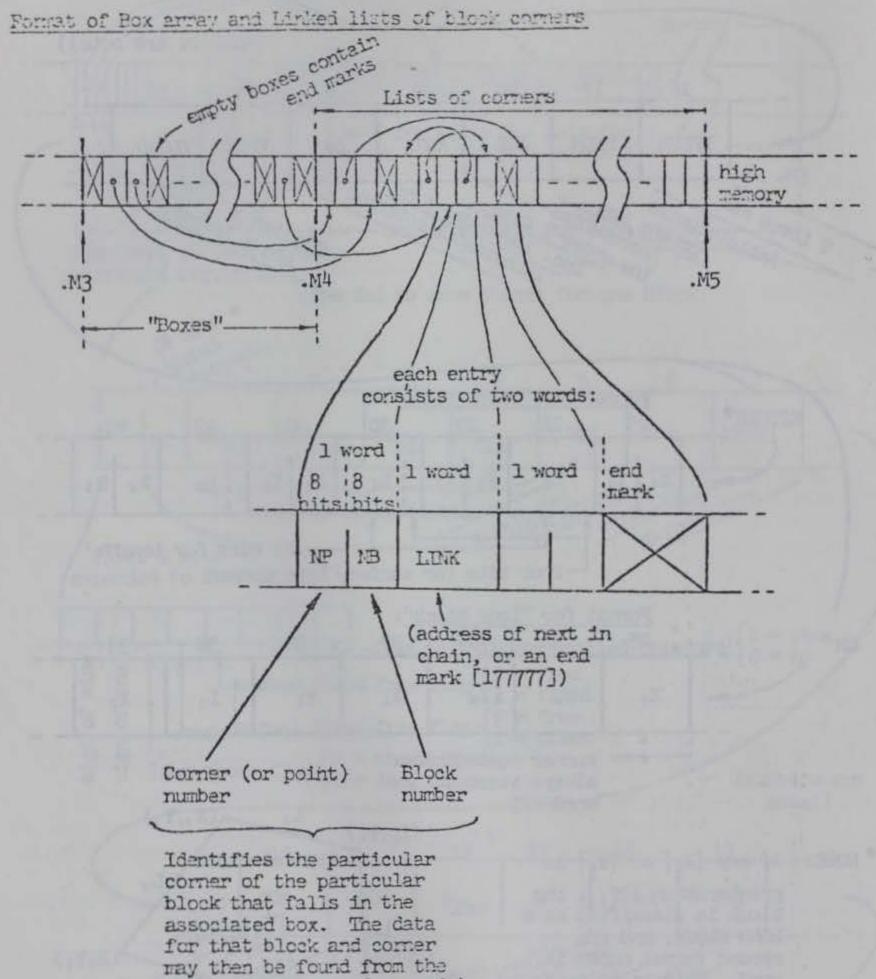
COS and SIN are stored as unsigned fractions (<1) (i.e. 1.0 is stored as 177777%)

continued on next page ...



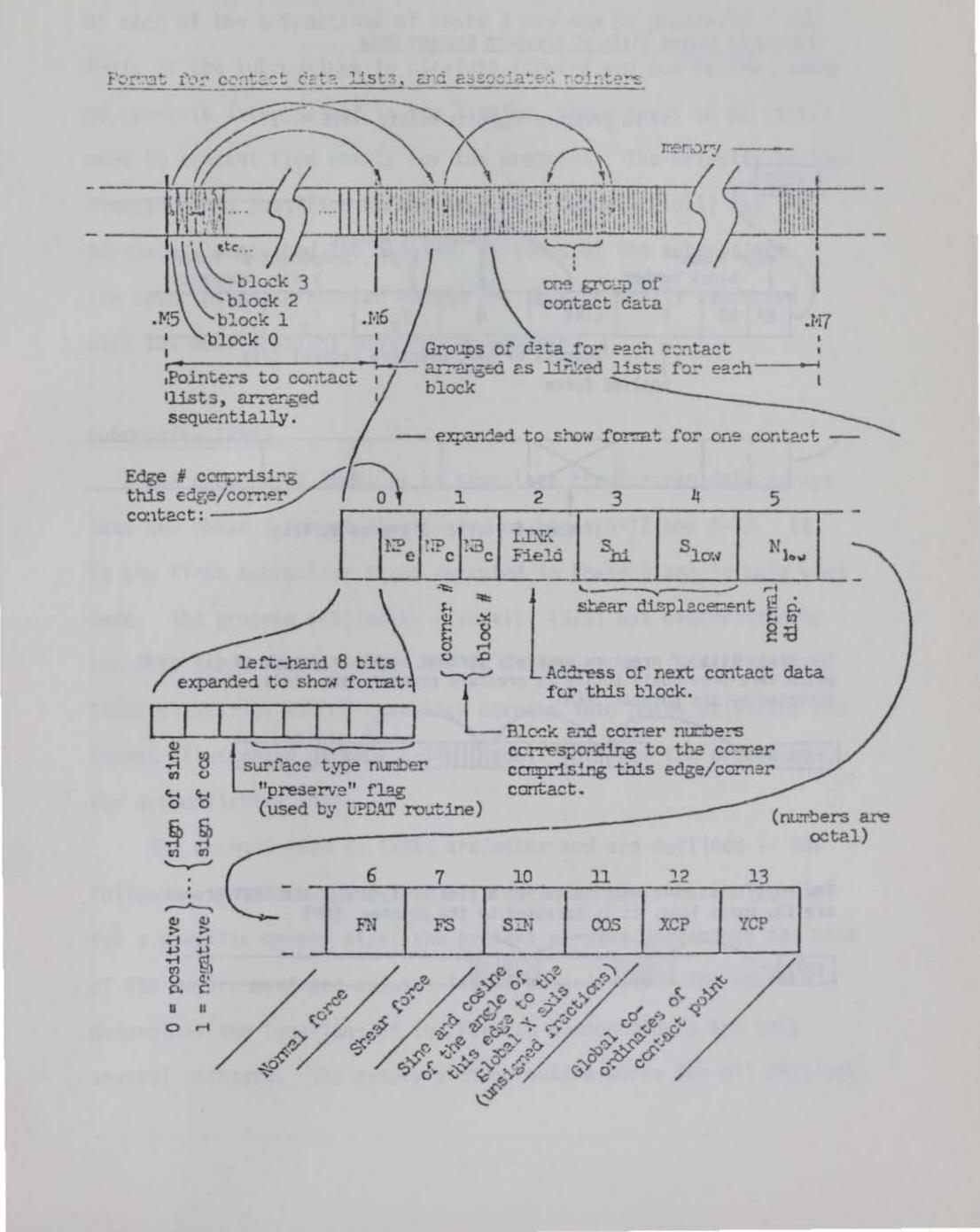
used. This is to save memory, as only a few blocks will be long.

 L_1 Numbering convention (X_2, Y_2) (Li are lengths) (X, Y,) are local co-ordinates



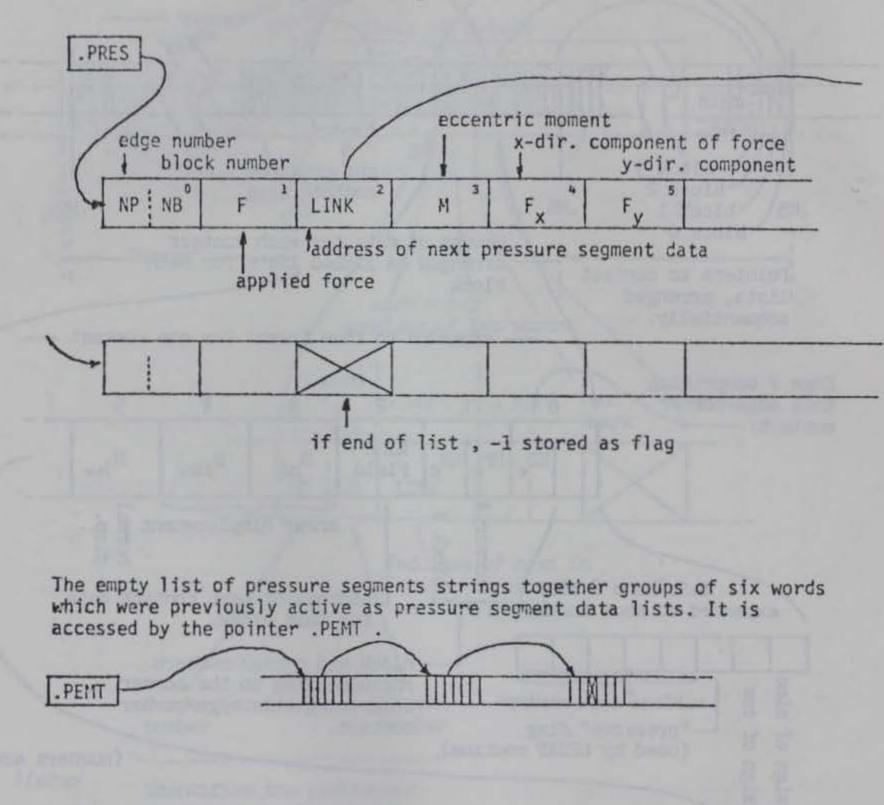
block data arrays (page 69)

Note: .M3, .M4 & .M5 are the global symbols (program names) for the pointers to the groups of memory shown



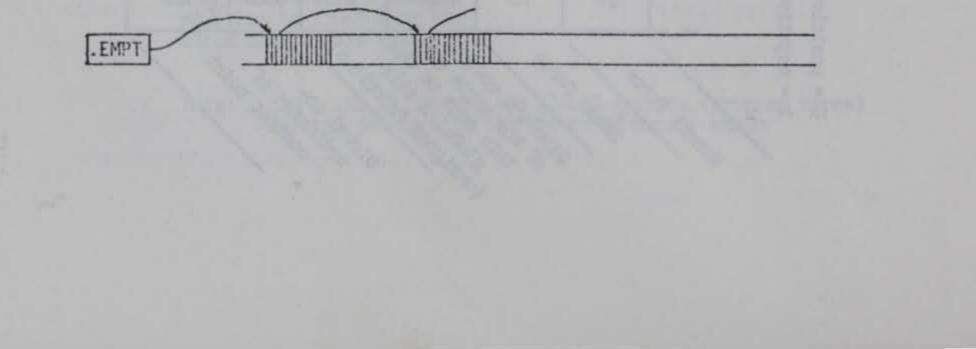
Format of Linked Lists of Pressure Segment Data

if no pressure segments exist, .PRES = -1



The empty list of contact data has a similar form but the list groups are 13, words long. It is accessed by the pointer .EMPT .

A-16



With this preliminary information in mind, a brief discussion of each of the subroutines of Phase 3 may now be presented. The logic of the subroutines is straight forward and due to the number of comments interspersed in the listing, there seems to be little need to present flow charts for the programs. The brievity of the discussion is justified by the fact that Cundall (1974) has adequately described the original versions of the subroutines. The descriptions presented herein are thus primarily concerned with the modifications made to the program.

Subroutine TRANS

The purpose of TRANS is to translate the Fortran data arrays into the Phase 3 format illustrated on pages A-12 and A-13. It is the first subroutine to be executed in Phase 3 and is only used once. The program originally (Cundall, 1974) was overwritten by the data input routine, but this is no longer so. Additionally, TRANS classifies all of the block corners into boxes utilizing the format illustrated on page A-14; Cundall outlines the procedure for accomplishing this.

The changes made to TRANS are minor and are outlined in the following sentences. The initial program version was implemented for a specific memory size; the present version determines the size of its environment and adjusts itself accordingly. The routine

determines the locations of the Fortran common blocks and sets

several pointers. The memory sizing routine works for all physical

configurations except 32K words; for this memory size the common block locations are displaced by one word. For this reason variable IY is dimensioned as 513 <u>only in Phase 3</u>. This juggling is not necessary for other memory sizes and may not be necessary for other operating software.

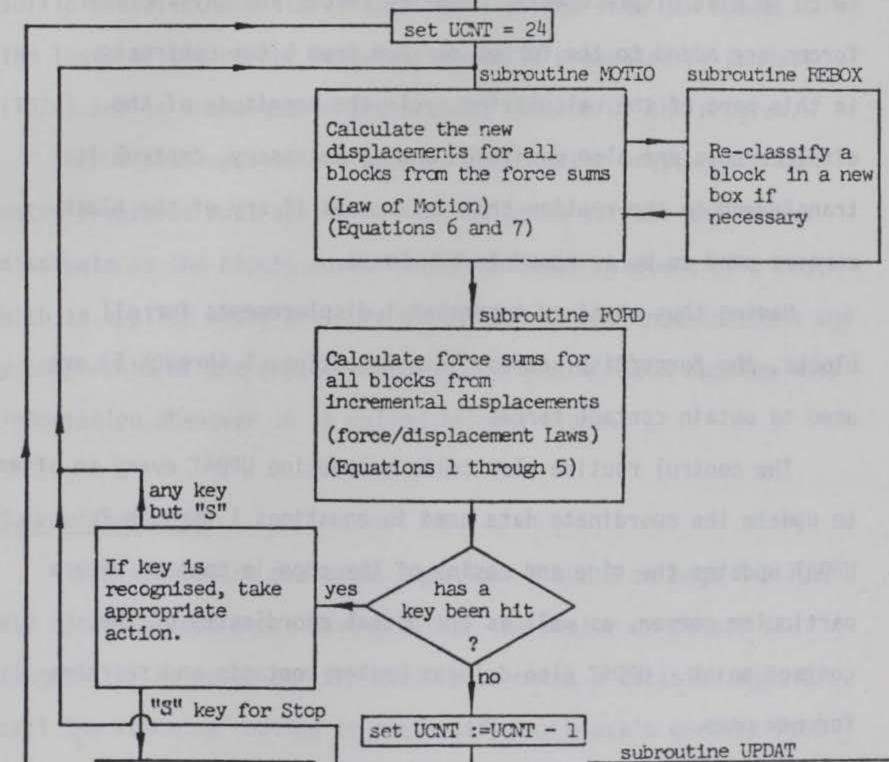
The veryone of many is to translate the formal and the sector and the sector of the se

. The country words to Table ore since and are outilized in the

Nor a second the second size of a present with a second state of the second second second second second second

Subroutine CONTR

The next routine to be executed governs the main control loop; subroutine CONTR also monitors the keyboard. The logic of the program is unchanged from Cundall (1974) but the fact that this routine embodies the main calculation cycle merits the presentation of a flow chart.



STOPPED For all blocks:update all Wait here for key; 15 yes existing contacts; UCNT = 0 no take appropriate search for and create new action if a key is contacts where necessary; hit. delete broken contacts. "G" key, for GO

The overall logic of CONTR is straight forward and simply involves the evaluation, for each block in turn, of the sets of equations listed on pages A-5 through A-7.

The calculation of the displacements from the forces (subroutine MOTIO) involves the evaluation of equations 6 and 7 for each block. Accelerations derived from forces are integrated twice to give displacements. Gravity forces and any applied forces are added to the forces derived from block contracts. In this part of the calculation cycle the magnitude of the displacements are also monitored and if necessary, control is transferred to the routine that determines if any of the block corners need to be assigned to new boxes.

Having thus obtained incremental displacements for all blocks, the force/displacement laws (equations 1 through 5) are used to obtain contact forces.

The control routine also calls subroutine UPDAT every so often to update the coordinate data used in equations 1 through 7. UPDAT updates the sine and cosine of the edge in contact with a particular corner, as well as the global coordinates of the contact point. UPDAT also deletes broken contacts and searches for new ones.

The other function of subroutine CONTR is to monitor the keyboard and respond to keys hit by the user while the program is running or waiting. The program responds to the keys and modifies the sequential operation of the program. The function of the individual keys is clearly explained in the listing of CONTR (Appendix C) as well as in Appendix B.

Subroutine REBOX

As has been observed, the corner reboxing routine is called from MOTIO whenever a block is suspected of having moved sufficiently to need its corners reclassified into new boxes. The logic of the corner reboxing scheme is presented by Cundall (1974) and is unchanged in the present version of the program.

REBOX also updates the applied joint water pressures. The water pressures must act normal to the joint surface and do not dissipate as the blocks move. Any rotational movement of a block with an applied water pressure would lead to a change in the x and y components of the applied force. Subroutine REBOX updates this information whenever it is called for any block.

Subroutine MOTIO

This subroutine evaluates equations 6 and 7 on page A-7 for all blocks except those having either the master or manual fix flags set. As noted earlier MOTIO also makes a decision when to call the reboxing routine to reclassify any block's corners into

new boxes. A call to REBOX is triggered whenever the cumulative

motion of any block exceeds one screen unit.

Laberaulter FORT is used in calculate the global chordinates of

Subroutine FORD

This subroutine evaluates equations 1 through 5 on page A-5 and A-6 for each block in sequence. It accesses the data stored in the contact list associated with each block, and computes the force sums acting on that block. Equation 5 is the only equation of the main calculation cycle that is different than that presented by Cundall. It now contains terms to account for the presence of joint water pressure.

 $F_{xsum}^{i} = \sum_{c} F_{x}^{ci} + F_{xload}^{i} + F_{xpres}^{i}$ $F_{ysum}^{i} = \sum_{c} F_{y}^{ci} + F_{yload}^{i} + F_{ypres}^{i} + F_{ygrav}^{i}$ $M_{sum}^{i} = \sum_{c} F_{y}^{ci} (x^{c} - x^{i}) - F_{x}^{ci} (y^{c} - y^{i}) + M_{pres}$ Ford also contains numerous entry points that are primarily used for experimenting with the program. These entry points allow modification of block weights and the dynamic factors of the program.

Subroutine UPDAT

The subroutine UPDAT is called once every few iteration cycles to check for new contact points. UPDAT also updates coordinate data as required. The routine is unchanged from the original form; the description presented by Cundall is very complete and contains a flow chart of the subroutine.

Subroutine PONT

Subroutine PONT is used to calculate the global coordinates of a contact point from the local coordinates of that point. This is

done by a simple coordinate transform for a translated origin and rotated axes. The equations are: (see any book on analytic geometry)

 $XG = XC + XL.cos\theta - YL.sin\theta$ $YG = YC + XL.sin\theta + YL.cos\theta$

where XL, YL = local coordinates

Subroutines DISPL and TEK

With the exception of the contact vectors, which are generated by subroutine FORD, all screen plotting is managed by subroutine DISPL. Subroutine DISPL in turn calls TEK which is nothing more than the basic Tektronix supplied software package for minicomputers. Whereas Cundall's (1974) version of the program provided hard copy through digital plotting, the present hardware includes a Tektronix 4631 copier. Although DISPL will still drive a digital plotter, this feature is rarely used.

The remainder of the subroutines of Phase 3 are primarily used for various utility functions. No great detail will be expended on describing the main function of each routine. The subroutine listings (Appendix C) contain many comments that indicate how the functions are performed. The interested reader is directed to the listings.

Subroutine INPUT

The utility routines embodied in INPUT are primarily concerned with parameter specification and modification. Most significant of the functions are:

- set up or modify the values of the ten different friction properties used by the program
- 2) input of applied pressures
- 3) numerical input of applied loads
- 4) set up of displacement control routine

The input of pressure segments deserves further attention. The presence of water in a joint tends to exert a force against the joint surfaces. For a single joint surface:

$$(x_{2}, y_{2})$$

$$(x, y)$$

$$(x, y)$$

$$(x_{1}, y_{1})$$

$$(x_$$

F and M are calculated as soon as a pressure segment is defined and never varies with displacement. The x and y components of the force do vary with displacement and are updated in REBOX.

4) $F_x = F.sin \alpha$

 $F_y = -F.\cos \alpha$

The initial value of F_{χ} and F_{γ} is also calculated in REBOX.

Subroutine UTIL

Subroutine UTIL contains several utility programs. The entry points and their functions are:

- HITC a routine to determine which block has the centroid corresponding to given x and y coordinates.
- 2) .PRN1 output a single character to the teletype
- 3) .ALPH sets the Tektronix to alpha mode
- 4) .PAGE a routine to clear the Tektronix screen
- 5) .LENG a routine to return the length of side NP

of the block in question

- 6) .TYP a routine to return the surface type number of a given edge
- 7) .SCAL a routine to scale vector lengths
 - 8) .IPRN a binary to decimal conversion routine that prints a right justified integer in a given field length

9) .PRN2 - a routine to print a single character on the teletype - character is in ACØ

10) .MESS - a routine to print a message at a specific location on the screen

11) .AXIS - a routine to draw an axis with tick marks

12) .GETT - a routine to receive a character from the

teletype

13) .DBIN - a decimal to binary conversion routine

it to binary if it is

15) .WORD - a routine to get an alphanumeric string from the key board

Subroutine CYCLE

Subroutine CYCLE contains several additional utility routines. The entry points and their functions are:

- .KET a routine to set velocities to zero at a kinetic energy peak
- RSET a routine to set the iteration cycle counter to zero
- 3) OPTIN a routine to set options governing vector scale factors, automatic copy and automatic stop
- 4) .STEP a routine to step the iteration cycle counter5) .TPRN a routine to print elapsed cycles

Subroutine HITS

Subroutine HITS checks all sides of all of the blocks to determine which edge of which block the coordinates x and y fall



Subroutine LOADS

Subroutine LOADS allows all block weights to be multiplied or divided by an integer constant.

Subroutine MOVIT

The law of motion for displacement controlled blocks is embodied in subroutine MOVIT

Subroutine TAPE

Subroutine TAPE contains the standard Linc tape utilities. It also contains the coding for reading or writing save files in Phase 3, and performs the overlay to return to Phase 1.

APPENDIX B

B-1

USER MANUAL FOR DISTINCT ELEMENT PROGRAM

The information contained in this Appendix describes the operation of the configuration of the Distinct Element program used for this dissertation. The Appendix is arranged in such a way that each of the three operating phases is described in sequence, with comment interspersed as necessary. The comment following the third phase of the program is extensive and contains much information pertinent to the successful operation of the program.

During all three phases of operation the computer responds to user commands whenever a teletype key is struck. There are a lot of key commands to which the program will respond with appropriate action. Lists of these keys follow. Rather than memorizing the lists and attempting to implement them all at once, it is strongly suggested that the potential user familiarize himself first with those keys which are essential to the operation of the program. As the user becomes confident in the use of these keys through the running of simple examples, more keys can be added to his "working vocabulary".

Essential Keys

Phase 1 - 1, 2, E, P-2, rubout

Phase 2 - E, S, R, P-3

Phase 3 - G, D, F, C, Z, I (F), S If a more detailed introduction to the use of the program is desired

see Cundall (1974).

PHASE 1 - OPERATIVE KEYS, CURSOR DISPLAYED

- 1 Key "1" is always used to define the first end of a line segment. Move the cross-hair cursor to the desired point and strike the key. The computer responds by drawing a "+" at the point indicated.
- 2 Key "2" is always used to define the second end of a line segment. Move the cross-hair cursor to the desired point and strike the key. The computer responds by drawing a "+" at the indicated point and by drawing a line between the first and second end points of the desired line segment. The computer program was modified to recognize the fact that it is often desirable to draw connected line segments. Therefore, the program will respond to the "2" key following either a "1" key or a "2" key. In this case the program supplies the coordinates of the first endpoint of the line segment at the proper time by using the last input of the second end of a line segment.
- E Any individual line segment may be erased by placing the cross-hair cursor at any position on the line segment and typing the "E" key. A useful trick to make the drawing clearer is to create a line segment at the edge of the Tektronix screen and then erase it. When the remaining line

segments are redrawn, the "+'s" at the ends of line segments are not redrawn.

rubout key. When the "E" key is used to erase a line segment, the end points of that line are not removed from the point list. These points can often impede the creation of a drawing.

If a large number of line segments are to be erased, it is preferable to use the "rubout" key.

- H To make a hard copy of the Tektronix display type key "H" or strike the make copy button on the console.
- W(code) To store the complete list of line segments created in Phase 1, type "W" followed by the desired code file number. To store the line segments in the third file, for example type "W" followed by "3".
- R(code) To recover a list of line segments created at an earlier time, type "R" followed by the desired code file number. For example, to recover the eighth file type "R" followed by "8".

Note: The program uses the ASCII equivalent of the

character to calculate the position of the file on the Linc tape. On a 620s block tape the permitted files, in order, are: 1-9, :, ;, <, =, >, ?, @, and A - Q. The program also stores a "password" in the file to prevent garbage from being read into the program.

N - The program has a subroutine to allow the numerical input

of line segment end points. To implement this feature,

С

type key "N".
 The Tektronix screen coordinates are from 0 to 1023 in the x direction and from 0 to 780 in the y direction. Often,
 the pucklem to be applying on be in field coordinates

the problem to be analyzed can be in field coordinates

which do not fall conveniently in this range. By typing key "C", a scale factor may be input to the program which is then used by the program to divide the input data in such a way that it will fall within the range of the Tektronix screen coordinates. Incidentally, the program treats both the scale factor and the input data as integer numbers, so nothing is to be gained by typing in highly accurate field coordinate data. The "C" key does not affect either the cross-hair cursor input or the digitizer input.

The program contains a subroutine to allow input of data by D means of a graphic tablet or digitizer. To implement this feature type key "D".

DIGITIZING ROUTINE

The digitizing routine will accept input data from the graphic tablet until the "E" key is typed. At this point the control returns to the main program and the cross-hair cursor is displayed. NUMERIC INPUT ROUTINE

Upon entrance to the numeric input routine, the computer responds by typing "Xl=?" and waiting for input data. After the data input following "Y2=?" several keys are operative.

- striking the carriage return key causes the computer to CR respond "X1=?" etc.
 - striking the "/" key causes the program to use the last endpoint as the first endpoint of a new line segment. The computer response is thus "X2=?" etc.

- L striking the "L" key causes the computer to redraw all lines. This key is frequently used as every input data pair will leave "X1=?" and "Y1=?" typed on the screen - it soon becomes difficult to follow what is happening on the screen unless "L" is frequently implemented.
- E striking key "E" while in the numeric input routine will cause control to be returned to the main program and the cursor is displayed.

Once the desired number of line segments has been created, the second Overlay of the program may be implemented. To do this, strike key "P" followed by key "2". Two comments are appropriate. First, it is not possible to get to Phase 2 from either the numeric input routine or the digitizer routine. The cross-hair cursor must be displayed before control can be passed to Phase 2. Second, all three input methods work together. Thus, it is possible to create part of the assemblage of line segments in the numeric input routine and finish the creation in the cross-hair cursor input routine.

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PHASE 1 SUMMARY

A) Cursor Displayed - Operative Keys

- 1 Use the cursor position as end no. 1 of a new line
- 2 Use the cursor position as end no. 2 of new line (display the line)
- E Erase the indicated line
- H Make a hard copy of display

rubout - Erase all lines

- W(code) Write the display onto tape in location code
 - R(code) Read the display at location code into memory
 - D Go to digitizing routine
 - N Go to numeric input
 - C Change N scale factor
- P Then 2 go to P-2
- assa vand införation of the second interest in the second second the second sec
- B) Digitizing Routine

 - Accept line segments from digitizer
 - E Escape to cursor on
 - upon entroyed in the accord a week boat you, the por
- C) Numeric Input Routine

1

- Responds X1=?, etc, after Y2=? several keys are operative:
- CR Select a new point

- on beleec a new porne
 - Repeat point
 - L Redraw all lines
 - E Escape to cursor on

PHASE 2 - OPERATIVE KEYS

- E A single block may be erased in Phase 2. To implement this option, place the cross-hair cursor on the desired block centroid and type key "E".
- R All erased blocks may be restored by typing key "R".
- A single block may be examined by placing the cross-hair cursor on the desired block centroid and typing key "S".
 After the single block is displayed, the block may be erased by typing key "E". Striking any other key returns without erasing the block. This feature is most useful to determine which centroid belongs to a given block.
- A Striking key "A" will display all of the blocks.
- H A hard copy of the display may be obtained by striking key
 "H" or pressing the "make copy" switch on the Tektronix console.

To return to Phase 1, strike key "P" followed by key "1".

To pass control to the third Overlay, Phase 3, type key "P" followed by key "3".

Two comments are in order. First, it is more economical in terms of computer work expended to erase unwanted blocks in Phase 2 than in Phase 3. Second, if the computer determines that no blocks can be created from the line segments passed by Phase 1, control is automatically returned to Phase 1. This means that it is not possible to get to Phase 3 without at least one block on the screen. To access a Phase 3 save file it is necessary to create a single

block, and pass it from Phase 1 to Phase 2 and then onto Phase 3.

At that point, the Phase 3 save file may be read. PHASE 2 SUMMARY Erase the block indicated E Display all blocks A S Display the single block indicated Erases the block, any E other key returns without erasing block Make a hard copy of the display H Restore all erased blocks R P then 1 go to Phase 1 P then 3 go to Phase 3

intention from the line constant for these of the second for the state of the state of the second for the secon

PHASE 3 - OPERATIVE KEYS

Iteration Cycle Not Running

- G To begin or continue the iteration cycle type key "G"
- D As the Tektronix is a storage CRT all images drawn on the screen remain on the screen until erased. To redisplay the system of blocks type key "D".
- Z To remove all inertia from the system type key "Z" to set all velocities to zero. This key is useful in the consolidation phase of the program in conjunction with the "V" key as described in a later section.
- H To make a hard copy of the blocks displayed on the screen type key "H" or depress the "make copy" switch on the Tektronix console.
- T To display the surface properly types which have been declared in the cursor routine, type key "T". The program displays a number from 1 to 9 at the midpoint of the edge of the block. Those surfaces having surface type Ø (the default value) are not indicated.
- W To store page zero (a variable list) and all block data, type key "W". The program writes this data on Linc tapes for future retrieval. This feature can be used to store the consolidated

block assemblage and identical problems can be run to study
 the effect of certain parameters. Only one file can be
 written or read by Phase 3, so no "code" is required.
 R - To read a previously stored Phase 3 write file, type "R". The
 program reads page zero and the block data, essentially

defining a new problem. A problem may be written on tape and returned to at a later time. As noted earlier, it is not possible to gain access to Phase 3 without going through Phase 1 and Phase 2. The best method of access is to create a single block in Phase 1 and pass it on to Phase 3. Upon typing key "R", the stored problem will be recovered. It is important to note that only the default friction value is stored in page zero. Friction properties for surface types 1 - 9 must be re-entered if the problem is changed. Note that it is possible to use the Linc tape utility "KBEX" to go directly to Phase 3, but this requires knowledge of several starting addresses.

- The contact vectors of each block may be displayed by typing V key "V". The stability of a block can be assessed by repeatedly typing key "V" and noting the variation of the position and length of the contact vectors. Note, however, that while the iteration cycle is not running, new contacts are not being detected (subroutine UPDATE) and repeated typing of key "V" may allow blocks to punch through edges. It is recommended that no more than 10 "V" keys by typed without typing key "G".
- The weights of all blocks, all externally applied loads and

- - joint fluid pressures are displayed when key "L" is depressed.
- To input joint fluid pressures, type key "J". The program J
 - responds by displaying the cross-hair cursor and waiting.

Position the cross-hair cursor on the desired joint segment and type the desired value of pressure followed by a carriage return. The cursor is then re-displayed. Additional pressure data may then be entered by the above procedure. Alternatively, a carriage return exists from the routine. Note that if two line segments are adjacent the logic of the program will apply to fluid pressure to both surfaces.

- C Typing key "C" displays the cross-hair cursor and allows entry to several input routines described in a later section.
- I By typing key "I", four additional input routines may be accessed by typing an additional key. These keys are:

F - If key "F" is typed following key "I", the routine to define surface friction property types is accessed. To define the friction coefficient corresponding to each numbered surface type, place the horizontal cursor on the same line as the desired surface type, type the "." key followed by a 3 digit decimal value of the friction coefficient, and end with a carriage return. After all desired friction coefficients have been defined, another carraige return will give control back to the main routine. Note that the maximum friction coefficient is 0.999 and that the value actually used by the program differs by .001 due to a validity check.

L - Typing key "L" following key "I" accesses the same numerical input routine described under key "O" in the cursor routine.

0 - Typing key "O" following key "I" allows the user to define several options including the options to print values of applied loads and contact vectors, define the vector length scale factor, and automatically make copies and stop the program after a desired interval. The kinetic energy damping routine should be used with extreme caution.

- U If key "U" is typed following key "I", a routine to define user units is entered. At the present time the only result of entering this routing is to cause a set of divided axes, labeled in desired units to be displayed on the screen.
- X By typing key "X" the iteration cycle counter is reset to zero. This routine is useful to set the cycle counter to zero after the consolidation phase so that the problem can begin at zero time.
- Q Typing key "Q" accesses several routines to vary some of the dynamic parameters and block weights. Its primary function is in program development and debugging.
- M Typing key "M" puts the cross-hair cursor on the screen and

enables the selection of the block to be used for the displacement control mechanism. Place the cursor on the desired block centroid and hit any key except "E". The program guides the user through the specification of the displacement steps, frequency and direction. Striking key "E" disables the mechanism if it is already set.

P - Upon completion of the problem, control may be passed to Phase 1 by typing key "P".

Iteration Cycle Running

- S To stop the iteration cycle and prepare for input, modification etc. type key "S".
- N While the iteration cycle is running blocks that are moving are being redrawn as they move. To prevent this type key "N". The computer responds by blanking the Tektronix screen. This action is required if the program is to be left unattended as the Tektronix screen can be permanently damaged if an image is displayed for a time longer than about 15 minutes without being redrawn. This option also makes the program run faster since the computer does not have to service the Tektronix for plotting.
- A Plotting of the blocks as they move can be restored by typing key "A". However, this option does not redraw all of the blocks, it only enables the drawing of blocks as they move.
 This has the advantage of allowing the user to determine

zones of movement within a mass, for example. To redraw all of the blocks, both moving and stable, type key "A" followed by key "D".

Several of the keys which are operative when iteration cycle is stopped are also operative when the iteration cycle is running. These are:

- D display all blocks
- H make a hard copy
 - display surface types
 - V display contact vectors
 - display load vectors

millight and

Iteration Cycle not Running, Cross-Hair Cursor Displayed
F - To force the program to hold a block fixed in space, place
the cross-hair cursor on the desired block centroid and type
key "F".

- U To release the status of a previously fixed block, place the cross-hair cursor on the desired block centroid and type key "U".
- E Blocks can be erased by placing the cross-hair cursor on the desired block centroid and typing key "E". However, as mentioned earlier, it is more economical in terms of computer effort to erase blocks while in Phase 2.
- O Typing key "O" writes the prompt message "Select Single Block". Place the cross-hair cursor on the desired block, hit any key and the program displays just the one block. Also displayed

on the screen are the block centroid coordinates and the magnitude of the applied loads. Additionally, if switch zero on the computer console is in the up position, pertinent force and velocity data are displayed. Finally, an opportunity is presented to numerically change the values of the applied loads. This routine exits the cursor routine automatically.

- 1 Applied loads may be input from the cursor routine by placing the cursor on the desired block centroid and typing key "1". The cross-hair cursor is then moved to a position defining the magnitude and direction of the desired load vector and key "2" is typed.
- Ø-9 Surface property type flags are set in the cursor routineby placing the cross-hair cursor on the desired block edge and typing a key from "Ø" to "9". This flag alerts the program to search the friction table for a specific friction value. Any other key removes the cursor and transfers control back to iteration cycle not running status.

There are two external "flags" available to the user to modify the execution of the program. These are data switches on the console of the computer. If switch 15 is in the up or on position, the printing of the elapsed cycles and default friction coefficient is inhibited. This is of use when it is desired to have copies that are free of text. The other flag is controlled by switch \emptyset on the console; it serves multiple purposes in guiding program execution. If switch \emptyset is in the up position, it is not possible

to return to Phase 1; this is done to prevent accidental loss of a program. Switch \emptyset "on" also causes velocity and acceleration data to be printed when a single block is examined, as well as allowing a message to be printed when the displacement control mechanism is operative.

21-12

PHASE 3 SUMMARY

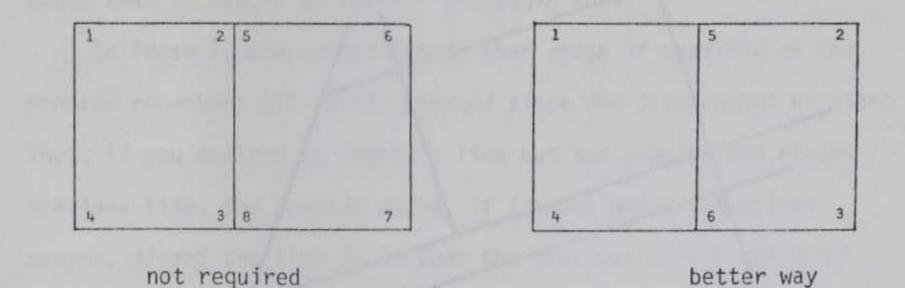
No	ot Running	Running
G	Go (start dynamics)	S Stop running
D	Redraw all blocks	N No plot option
Ζ	Set all velocities to zero	A Activate plotting
Н	Make hard copy	Also: D, H, T, V, L
т	Display surface types	
W	Write display on tape	Cursor Displayed
R	Read display from tape	F Fix block indicated
۷	Display contact vectors	U Unfix indicated block
L	Display loads & pressures	E Erase block indicated
J	Accept joint pressures	O Display block indicated
С	Display cursor	
I	Input activation	load vector (centroid) followed by a 2
	F Friction U Units	
	L Loads O Options	type (friction)
Х	Reset cycles	Other keys remove cursor
Q	Debug routine	
Μ	Access displacement control	
Р	Go to Phase 1	

USEFUL INFORMATION

The remainder of this Appendix is devoted to the presentation of information that will be of use to potential users of the program. Some of this information is intended to make it easier for an untrained user to begin working with the program, some of it is intended to aid those interested in program development and some of it is simply odds and ends. No apology is offered for the rather rambling nature of the presentation.

Block creation

In the first overlay or main section of the program, line segments are drawn on the Tektronix screen using the cross-hair cursor, a numerical coordinate input routine or the graphic input tablet. At this stage of the program we are only drawing line segments. Thus it is not necessary to draw each block individually.



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The program detects intersections and overlaps and treats them as such. Incidentally the program has a built in error factor of 5 screen units (out of 1023 x or 768 y). It is therefore impossible

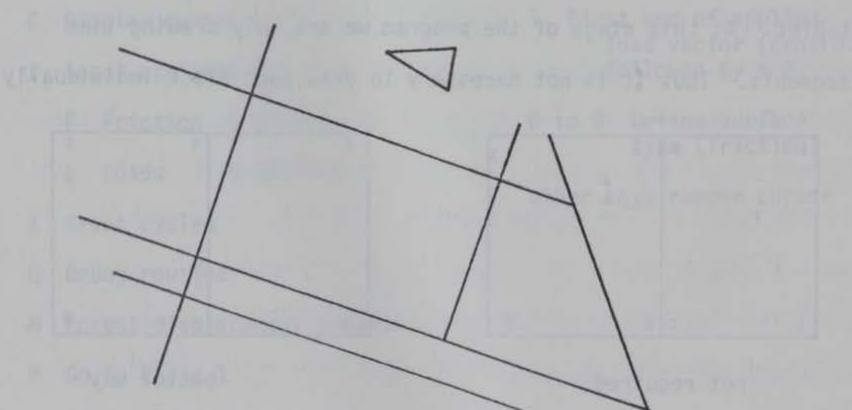
to create a situation such as:

< 5 units

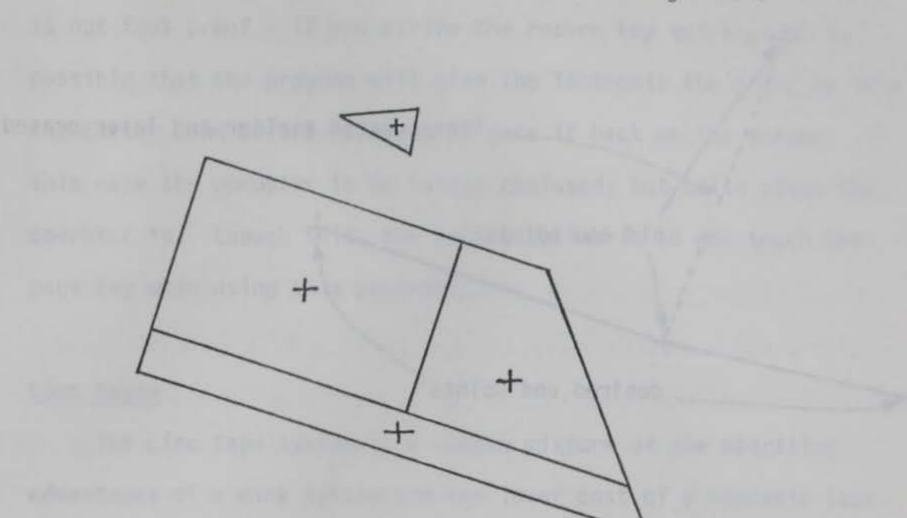
Sector Internet Parts Str.

The program will merge the points into

Always remember that line segments that do not define a closed area will be rejected by the program Overlay 2 (see following paragraph). In the second Overlay of the program, the computer scans all line segments created in the first Overlay to determine which line segments will form closed areas. For example, if the following line segments were created in Phase 1, (or the first Overlay):



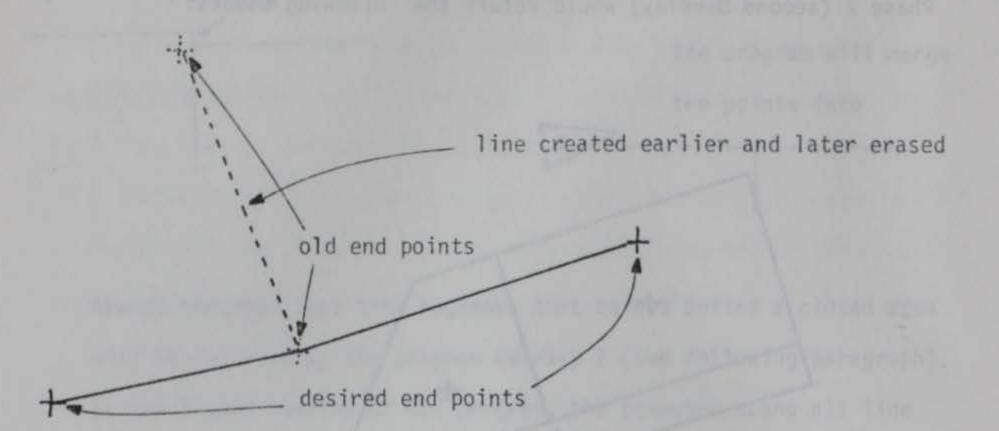
		1	
		1	



Phase 2 (second Overlay) would return the following blocks:

It must be emphasized that closed areas must be drawn in Phase 1 if blocks are desired in the main part of the program. If a desired line segment has been inadvertently omitted, there is no recourse other than to return to Phase 1 and begin anew.

In Phase 1, use rubout rather than erase if possible as the program remembers all points created since the last rubout or start. Thus, if you desired to create a line but had created and erased a previous line, the program would, if it considered the action proper, divert the line to include the previous line's end point.



This happens very easily, be aware of why it happens.

As the Tektronix 4010-1 is a storage oscilloscope and not a television screen, all information drawn on the screen is stored on the screen. Under no circumstances use the page key to clear the display. This leads to a minor state of confusion as to what the program is doing. Especially serious is the situation that occurs if you use the page key when the cross-hair cursor is displayed. The effect of this is to place the screen in ALPHA mode (ASCII input) while the governing software is still in GIN MODE

(graphic input). When this occurs, you no longer will be able to

communicate with the computer through the Tektronix, and the

computer will be hung-up in the graphic input loop. This isn't really as serious as it looks. For some reason, striking the

return key several times will bring the cursor back. However, this is not fool proof - if you strike the return key quickly, it is possible that the program will give the Tektronix the order to take the cursor down before it actually gets it back on the screen. In this case the computer is no longer confused, but quite often the operator is. Enough said, the best solution is to not touch the page key when using this program.

Linc tapes

The Linc tape system is a unique mixture of the operating advantages of a disk system and the lower cost of a magnetic tape format. The addresses of the storage blocks are written on the tape and the software can search the tapes in either direction for a specific block address and, once it is found, read, write or overwrite starting at that address. The present form of the Distinct Element program relies heavily on the Linc tapes and the following paragraphs present information that could be of use to someone using the program.

The system used for this study has two drives - unit 0 and unit 1. Unit 0 is used by the program for the Phase 1 save files. The save file handling routine, subroutine TAPE, does not check the

tape file directory before writing nor does it append a title to the

directory for the save file. It is thus a good idea to use a blank

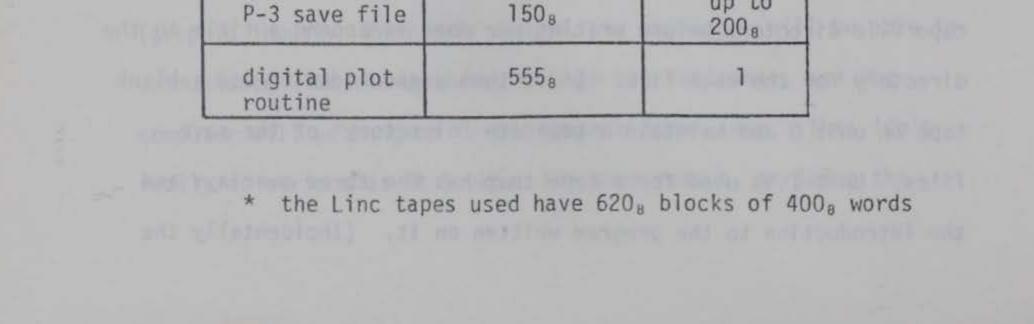
tape on unit O and maintain a separate "directory" of the save

files. Unit 1 is used for a tape that has the three overlays and the introduction to the program written on it. (Incidentally the program is assessed by placing a "blank" tape on unit O, a "program" tape on unit 1 and typing "HELP". The program takes it from there!) The tape on unit 1 is also used to store the Phase 3 save file. It is important to note that the file directories do not "know" about the overlays and save file and thus it is up to the user to protect all file space from block 150_8 onward.

The Linc tape furnished software used in this study did not possess a sophisticated operating system. The fact that not having a sophisticated operating system led to additional memory (= larger problems) was offset by the fact that the overlays must be "done by hand".

The Linc tape utilities have the capability to move data from the tape to memory and vice versa. The overlays of the program are simply immages of memory written onto tape. For the present study the pertinent addresses on the tape on unit 1 are:

tape file	beginning block number *	number of blocks	
Phase 1	3508	55 ₈	
Phase 2	450 ₈	378	
Phase 3	5108	378	
		up to	



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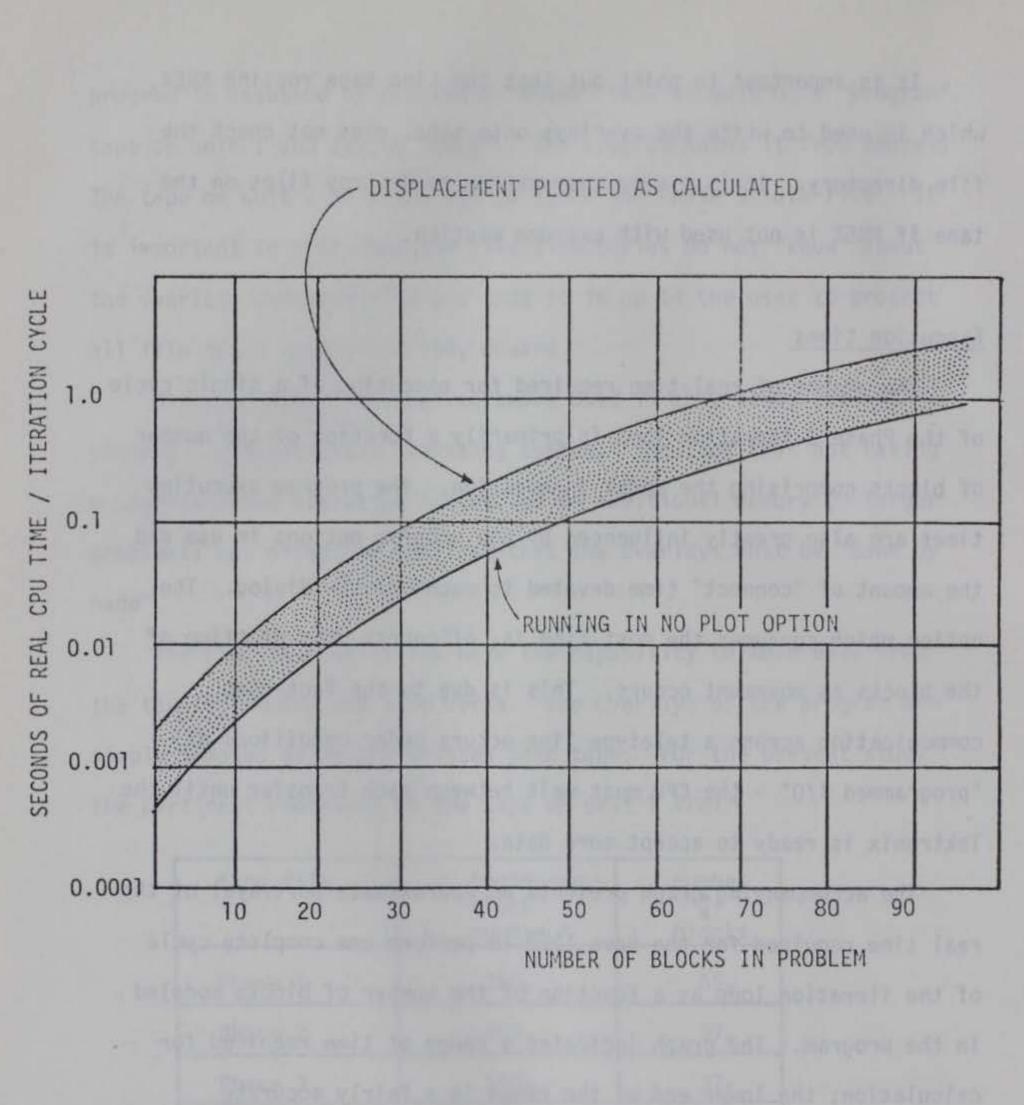
It is important to point out that the Linc tape routine KBEX, which is used to write the overlays onto tape, does not check the file directory. It is a very easy matter to destroy files on the tape if KBEX is not used with extreme caution.

Execution times

The amount of real time required for execution of a single cycle of the Phase 3 iteration loop is primarily a function of the number of blocks comprising the model in question. The program execution times are also greatly influenced by any program options in use and the amount of "connect" time devoted to machine/user dialog. The option which consumes the most time is, of course, the plotting of the blocks as movement occurs. This is due to the fact that communication across a teletype line occurs under conditions of "programmed I/O" - the CPU must wait between each transfer until the Tektronix is ready to accept more data.

The accompanying graph presents an approximate portrayal of the real time required for the Nova 1220 to perform one complete cycle of the iteration loop as a function of the number of blocks modeled in the program. The graph indicates a range of time required for calculation; the lower end of the range is a fairly accurate

representation of the fastest possible calculation times for a given number of blocks. This time can only be realized by running in the "no plot" option. The upper end of the range represents the time required for one cycle of the iteration loop with the plotting option



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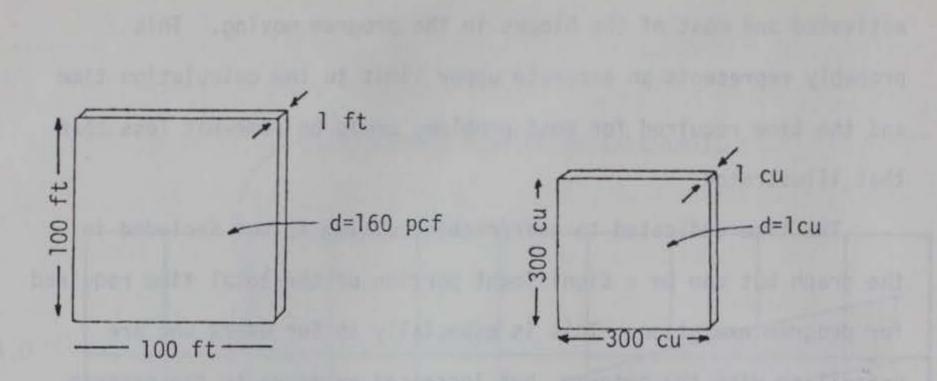
activated and most of the blocks in the program moving. This probably represents an accurate upper limit to the calculation time and the time required for most problems would be somewhat less than that illustrated.

The time dedicated to user/machine dialog is not included in the graph but can be a significant portion of the total time required for program execution. This is especially so for users who are unfamiliar with the program, but increased exposure to the program usually leads to familiarity and an attendant drop in the amount of time required for interaction.

Conversion factors

All calculations performed by the Distinct Element program described in this Appendix utilize variables whose magnitudes and dimensions have been adjusted to give optimum calculation speeds. This has been done in order that double precision variables are avoided and so that all arithmetic is done on integers (integer arithmetic is many times faster than floating point arithmetic in the absence of a floating point processor). In order that someone who wishes to do so may convert to either metric or english units, three conversion factors are presented in the following paragraphs.

The first conversion factor is a <u>defined</u> relationship between physical problem length and that used in the computer program. Consider the following physical situation: a block 100 ft on a side, 1 ft thick, with a unit weight of 160 pct.



The computer model is drawn in such a way that the equivalent edge lengths are 300 cu (computer units). The unit weight in the computer model is 1 cu (this can be changed by typing "Q" followed by key "W" - the following must be modified if the unit weight is changed). By selecting 300 cu to represent 100 ft, the first conversion factor f_d is automatically defined.

To get feet or meters multiply the program distance by fd

In this particular example,

 $300 \text{ cu} + f_d = 100 \text{ ft}$ or $f_{d} = 0.333 \text{ ft/cu}$

The second conversion factor is a derived relationship between physical problem forces and those used internally in the computer program returning to the example, the real weight of the block is

seen to be:

100 ft * 100 ft * 1 ft * 160 pcf = 1.6 x 10⁶ lbs The weight of the block in computer units is given by the Distinct Element program - in this case it is seen to be 720 cu. The number 720 represents a normalized weight obtained by determining the volume of the block and dividing by 125. The number 125 is related to the tolerance to which points and lines are subjected in Phase 1 and Phase 2. The smallest block allowed is defined to be 5 times the area defined by the screen accuracy (5 x 5). The smallest block area possible is then 125 units; when normalized the smallest block weight allowable is thus 1 cu since the unit weight used in the program is 1 cu. The weight used in the computer program for this example is thus

 $\frac{1}{125} * \frac{100 \text{ ft}}{f_d} * \frac{100 \text{ ft}}{f_d} * \frac{160}{d} \text{ pcf} = W \text{ cu/unit depth}$ Since W real/unit depth = 100 ft * 100 ft * 160 pcf
W real = 125 * f_d^2 * d * W cu

The conversion factor between real situation force and that used internally by the computer is f_{i}

$f_{l} = 125 + f_{d}^{2} + d$

To get force in pounds or newtons multiply the displayed force by f_{i} .

In this particular example

$$f = 125 \times 0.333 \times 160$$
 or
 $f = 2222.22 \ 1b/cu$

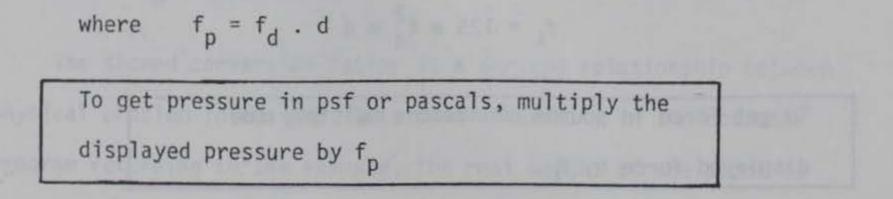
The third conversion factor relates pressure in physical units such as psf or N/m^2 to the units used internally in the computer program. If the base pressure of the real block considered in this example is calculated the quotient of the block weight and the contact area are found.

$$P_{real} = \frac{W}{A} = \frac{100 \text{ ft} \pm 100 \text{ ft} \pm 1 \text{ ft} \pm 160 \text{ pcf}}{100 \text{ ft} \pm 1 \text{ ft}}$$

In the computer situation this reduces to

P (cu) =
$$\frac{\frac{100 \text{ ft}}{f_d} * \frac{100 \text{ ft}}{f_d} * \frac{160 \text{ pcf}}{d} * \frac{1 \text{ ft}}{f_d}}{\frac{100 \text{ ft}}{f_d} * \frac{1 \text{ ft}}{f_d}}$$

or P real = P cu * fp



In the example considered, if it were desired to input a joint water pressure whose resultant would balance the weight of the block, its magnitude would be found in the following manner

- real pressure
$$P = 1.6 \times 10^6$$
 lb/100 ft² = 16000 psf

$$f_n = f_d \times d = 0.333 \times 160 - 53.3 \text{ psf/cu}$$

- pressure in computer units =
$$\frac{P}{f_p}$$
 = $\frac{16000}{53.3}$ = 300 cu

Equilibrium conditions

The problem of recognition of equilibrium conditions is of paramount importance in the Distinct Element method, as in other explicit finite difference programs. An explicit formulation does not have a "solution" in the sense that an implicit formulation such as a Finite Element analysis does. In the implicit formulation the behavior of each point is related to the other points through a system of equations that can be solved for a given input resulting in a solution. In an explicit formulation, on the other hand, the points communicate only with their nearest neighbors; the "solution" in this case does not necessarily need to be a situation of stable equilibrium. The only way that an equilibrium situation can be recognized is by observing the behavior of the blocks.

The obvious solution to this problem is to observe the blocks

flashing on the screen - the movement of the blocks is obvious and

it can immediately be recognized if the problem under consideration is unstable. However, the fact that the blocks are not flashing

on the screen does not necessarily indicate that an equilibrium situation has been reached. In the example considered in the previous section, one screen unit of displacement corresponded to four inches of real displacement. In a large problem where the blocks are somewhat confined, thousands of iteration cycles will be needed to get this much displacement; for a program involving 75 blocks the real time for this many calculations could take an hour. This is obviously not a very satisfactory method to determine if equilibrium exists.

The software necessary for more subtle solutions has been incorporated within the present version of the program. At any time during the running of a problem, the program may be stopped (key "S") and any block examined for pertinent data. By displaying the cursor (key "C") then typing key "O" will result in the message "SELECT ANY BLOCK" being displayed on the screen. By placing the cursor on the desired block centroid and striking any key a display of block data will be presented. This data includes: block centroid coordinates (four places to right of decimal point displayed); the unbalanced force sums acting on the block; the block velocities and angle of rotation; and, the values of user applied loads. By examining certain "key"blocks as the program runs it is a relatively

simple matter to determine if an equilibrium state has been reached.

Block consolidation

The block data passed onto Phase 3 from the first two overlays

contains information pertaining to individual blocks only. The

contact lists do not exist before the start of the program, so the blocks do not know that they have neighbors. When gravity is suddenly switched on, all of the blocks begin to move at once and as block interactions occur, the contact lists are developed. The way in which the block configuration is allowed to interact has a significant effect on the outcome of the program in those instances where a proper mass consolidation is not achieved. An improperly consolidated system of blocks can lead to a diverging solution; this can be recognized by the presence of wildly fluctuating contact forces that bear no relation to the block weights involved.

The blocks should be allowed to consolidate in an initial equilibrium position before the actual problem is run. This can usually be accomplished by the judicious placement of restraining blocks; these are subsequently removed to begin the actual problem. To actually consolidate the mass a good deal of time must be spent observing the behavior of the blocks and intervening to guide the program. Just switching gravity on without regard to consolidation of the blocks can easily lead to situations where pressure waves travel through the mass and prevent the blocks from reaching an equilibrium state.

Several bits of information are related in the following

sentences that should be helpful to potential users of the program. First of all it is very helpful to start the problem with all frictional properties set to zero (the program automatically does this unless the user changes the friction table). The first block interactions often involve high contact forces; if the friction TE-d

coefficients of the surfaces are other than zero, situations can arise whereby relatively large forces are "locked-in" only to be released when just the right contact occurs. By starting with a zero value of the friction coefficient, shear resistances do not develop along the joints and in conjunction with the velocity zeroing technique described below, the restrained system of blocks comes to equilibrium. At this point, the restraining blocks can be removed and the program allowed to run.

The technique of properly consolidating a system of blocks involves zeroing the block velocities at the correct time; the system of blocks cannot reach equilibrium unless all inertial effects are removed. It is possible to gain insight into the status of a block mass by examining the behavior of the contact vectors. The key "V" is used to display the contact forces whenever it is struck; this is accomplished by setting a plot flag, going once through the iteration cycle and then taking the flag down. This is especially useful if the program is in the stopped mode since the "V" key can be used to step through the iteration cycle incrementally. The variation in the length and angle of the contact vectors is indicative of the relative stability of the behavior. Well consolidated systems of blocks display little

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variation in length or inclination of the contact vectors. To achieve this state the user must examine the behavior of the system and zero the block velocities (key "Z") when the system is in an "average" state. An "average" state is exactly what it sounds like - the length of the contact vectors are approximately the average of the variation in length, and the inclination of the contact vectors is approximately midway between the extreme inclinations. This can rarely be achieved in one attempt, and the amount of time required to do it successfully increases with the degree of confinement of the problem (i.e., tunnel models are much more difficult to consolidate than slope models).

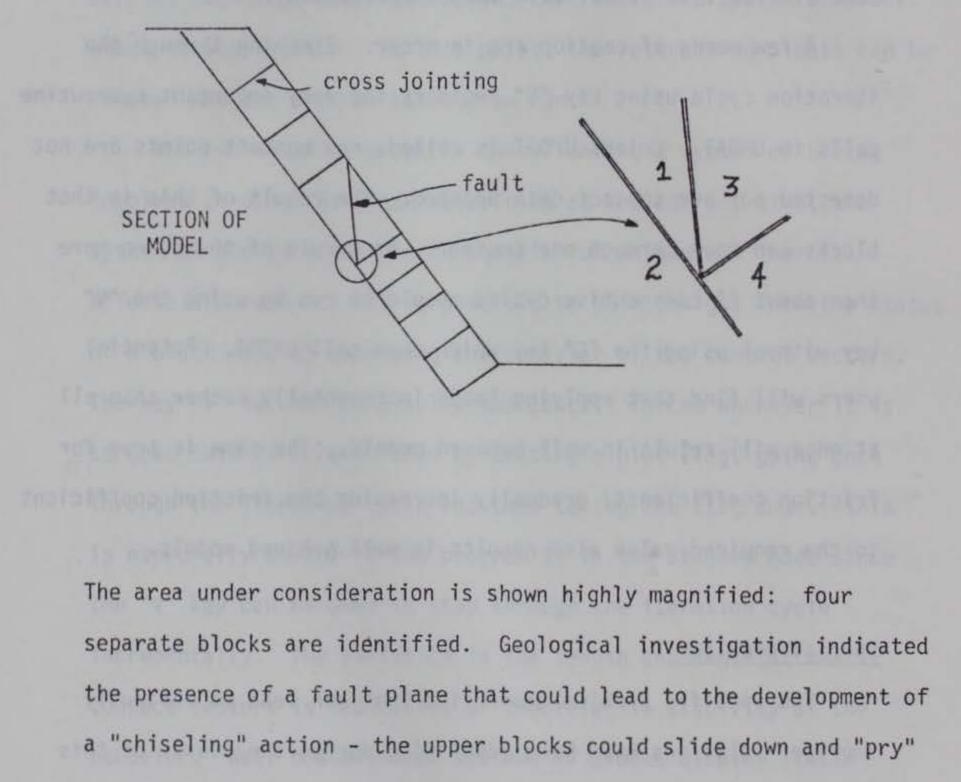
A few words of caution are in order. Stepping through the iteration cycle using key "V" neglects the very important subroutine calls to UPDAT. Unless UPDAT is called, new contact points are not detected nor are contact data updated. The result of this is that blocks can move through one another. As a rule of thumb, no more than about 25 consecutive cycles should be run by using the "V" key without using the "G" key which does call UPDAT. Potential users will find that applying loads incrementally rather than all at once will result in well behaved models. The same is true for friction coefficients; gradually increasing the friction coefficient to the required value also results in well behaved models.

Special problems

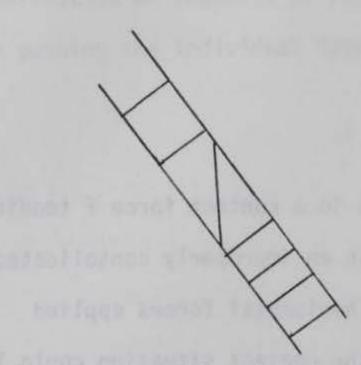
Two specific problem geometries that can lead to obviously improper solutions have been identified during the course of this

research. Both involve shortcomings in the contact determining logic; the problems are identical in nature but whereas one is easily overcome, the other requires that some care be expended in block consolidation to prevent its occurance. The problems will be illustrated by reference to the specific geometries in which they were first identified.

The first of the two problems occurred during the analysis of a rock slope which had failed. (This incidentally, was a real problem - the analysis was performed in collaboration with Dr. Michael Bukovansky of the consulting firm of Dames & Moore.) The geometry of the problem:

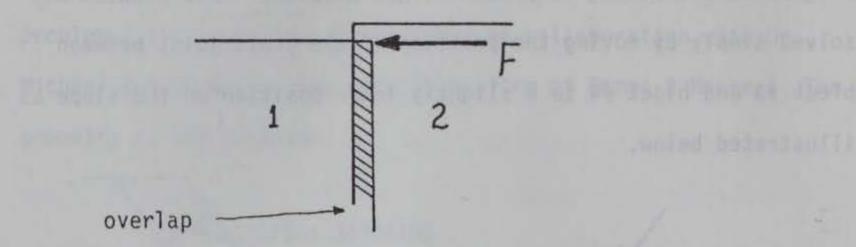


the lower blocks. The initial analyses performed using the Distinct Element program failed to reproduce the expected failure. Close examination of the behavior indicated that instead of sliding past block #3, the lower point of block #1 was contacting block #4 and "hanging up"; the net result being that the entire assemblage of blocks stabilized. In the real situation, any such contact would result in fracture development at the point - in the Distinct Element program such cracking is presently not modeled. This problem was solved simply by moving the position of the cross joint between block #3 and block #4 to a slightly lower position on the slope as illustrated below.



The second problem is of a similar nature; its occurance is rare and is usually due to improper block consolidation. The problem was identified in a model similar to that illustrated and resulted in the stability of a model which should have failed.

To illustrate the problem a magnified section of the model is required; a contact between blocks #1 and #2, circled in the sketch, is illustrated



The overlap of the two blocks results in a contact force F tending to push the blocks apart. However, in an improperly consolidated block mass, especially one with high horizontal forces applied before the mass is allowed to move, the contact situation could look like this after the first iteration.

Depending upon which "contact" is first discovered by the contact seeking logic edge #1 of block #1 could be identified as the edge in contact. The resultant force would thus act to prevent the downward movement of block #2. This problem has not arisen in models where proper consolidation steps have been taken. As

2

overlap

1

insurance, however, all models tested where this problem could occur have been allowed to fail as part of the analyses, to make certain that the problem was not occurring.

For those geometries to be tested where the occurance of this problem is a possibility, special care can be taken during the consolidation phase to prevent its occurance. This often involves consolidation of segments of the model on an individual basis and then pushing the individual segments together to form the model.

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APPENDIX C

LISTING OF THE DISTINCT ELEMENT PROGRAM

This Appendix contains listings of all of the subroutines necessary to build the three overlays of the Distinct Element program used in this dissertation. Most of the Phase 1 and Phase 2 routines are written in Fortran; a few are written in Data General Nova assembly language. All of the Phase 3 subroutines are written in Nova assembly language.

At first glance, the assembly language subroutines may appear to be of little value to those unfamiliar with Data General computers; this is, however, not the case. Assembly language programming differs very little from the techniques used in programable calculators and in fact rarely involves anything more sophisticated than moving data between memory and accumulators, performing arithmetic functions, and occasionally jumping to a subroutine. The listings presented are interspersed with numerous comments and the straightforward logic of the program makes them very readable.

As an aid to potential users a list of the subroutines loaded in each overlay is presented next.

List of Phas	e 1 Subroutines	Page Number
MAIN LINEX ERASE INSEC HARD CROSS		C-4 C-10 C-11 C-12 C-14 C-14
TEK TAPE COPY OVERLAP DIGIT	machine language subroutines; Fortran interface recognized by calls to .CYPL and .FRET.	C-15 C-19 C-23 C-24 C-27
List of Phas	e 2 Subroutines	Page Number
BUILD CENT CROSS HARD		C-29 C-33 C-14 C-14
TAPE COPY TEK	machine language subroutines; Fortran interface recognized by calls to .CYPL and .FRET.	C-19 C-23 C-15
List of Phase	e 3 Subroutines	Page Number
TRANS TEK PONT HITS TAPE UTIL LOADS FORD UPDAT	see note following	C-40 C-48 C-51 C-54 C-59 C-64 C-75 C-79 C-94

19-2

CONTR CYCLE INPUT MOVIT

REBOX

MOTIO

DISPL

C-120 C-138 C-149 C-166

C-104

C-108

C-113

Note The order in which the subroutines are loaded is immaterial unless the digital plotting routine (subroutine PLOT, Cundall, 1974) is desired. In this case, the plotting routine is read from the

tape, in absolute binary, whenever it is needed. The routine starts at location 440_8 and thus overwrites the first subroutine in memory. If the loading sequence places TRANS at the start of memory, the overwriting will not disrupt the program.

Preceeding the listing of the Phase 3 subroutines is a list of the Phase 3 global symbols. These are primarily entry point addresses and frequently used variables. The listing begins on Page C-37.

001	CMAI	IN PROGRAM (OVERLAY NUMBER ONE)
802		COMMON I1(768), 12(768), LISI(32),
003	*	LISTC(128), 1X(512), 1Y(512)
004		COMMON/HANDY/N,L, IACC
005	75	N=Ø
006		L=0
007		1400-5
008		IFACT=1
009	1	SXL=XLM
010		SYL=JYL
011		LCODE=0
012		KODE=0
013		CALL CURS(I, JX1, JY1)
014		CALL CHARO(159)
015		IF (N.EQ.0 . OR. I.NE.178) GO TO 80
016		LCODE=1
917		JX2=JX1
018		JY2=JY1
019		JX1=MJX
020		JY1=MJY
021		GO TO 103
022	80	IF(I.NE.196) GO TO 400 "D" FOR DIGITIZER
023	0.5	KODE=1
024		GO TO 100
025	400	IF(1.EQ.195) GO TO 210 ;"C" TO CHANGE FACTOR
026		IF(1.NE.206) GO TO 104 IN FOR NUM. INPUT
027		KODE=-1
028		GO TO 201
029	104	IF(1.E0.200) GO TO 72 ;"H" FOR HARD COPY
030		IF(1.EQ.197) GOTO 73 ;"E" FOR ERASE
031		IF(1.EC.208) GOTO 76 J"P" FOR "PHASE"
032		IF(1.EC.255)GOTO 74 ; RUBOUT ALL LINES
033		IF(1.E0.215) GO TO 81 ;""W" FOR WRITE
034		IF(I.NE.210) GO TO ET JMUST BE "R" TO READ
035		CALL CHARI(I)
036		NFIRST=(1-177)*12 JGET FILE CODE
037		CALL CHARO(155)
038		CALL CHARO(140)
039	83	CALL TAPE(1,NFIRST, 11, 11, NERR)
040	05	IF (NERR.EC.0) GO TO 82
041		PAUSE TAPE ERROR HIT ANY KEY TO REPEAT
042		GO TO 83
042	82	N=LIST(1)
043	08	L=LIST(2)
045		IF(LIST(3) • NE • 13286) GO TO 75
046		DO 84 LX=1.L
040		IA=I1(LX)
047		IB=I2(LX)
040		CALL PLOTS(0,1X(IA),IY(IA))
050	84	CALL PLOTS(1, IX(IB), IY(IB))
050	0.4	

051 CALL CHARO(159) GO TO 1 052 81 CALL CHARI(I) 053 NFIRST=(1-177)*12 054 955 LIST(1)=N

056		LIST(2)=L		
057		LIST(3)=13286		
058	86	CALL TAPE (2, NFIRST, 11, 11, NERR)		
059		IF (NERR.EQ.0) GO TO 1		
060		PAUSE TAPE ERROR WRITE PROTECT ON ? HI	T A KE	Y
061		GO TO 86		
062	87	IF(I.NE.177) GOTO 1 J"1" FOR FIRST EN	ND OF	LINE
063		IF (KODE . EQ . 0) GO TO 103		
064	100	CALL DIGIT(JX1, JY1, ICODE)		
065		IF(ICODE .NE .0) GO TO 1		
966		GO TO 103		
Ø67	201	ACCEPT" X1=", JX1," Y1= ", JY1		
068	201	JX1=JX1/IFACT		
069		JY1=JY1/IFACT		
070	103	IF(N.EQ.0) GO TO 4		
071	105	DO 2 NN=1+N		
072		IF(IABS(IX(NN)-JX1).GT.IACC) GOTO 2		
073		IF(IARS(IY(NN)-JY1).GT.IACC) GOIO 2		
074		IFIRST=NN		
075		GOTO 3		
076	2	CONTINUE		
077		GOTO 4		
078	3	JX1=1X(IFIRST)		
079		JY1=IY(IFIRST)		
080		IF(LCODE .EQ. 1) GO TO 108		
081		CALL CHARO(135)		
082		IF (KODE) 202, 14, 109		
083	4	IF(L.EQ.0) GOTO 12		
084	-	CALL LINEX(JX1, JY1, IXR, IYR, NHII, LL)		
085		IF (NHII.EQ.1) GO TO 8		
086	12	IFIRST=N+1		
087	10	GOTO 13		
088	8	JY1=IYR		
089	0	JX1=IXR		
090		IFIRST=N+1		
091		L=L+1		
092		II(L)=IFIRST		
093		12(L)=12(LL)		
094		I2(LL)=IFIRST		
095		CALL CHARO(135)		
096	13	IX(IFIRST)=JX1		
097		IY(IFIRST)=JY1		
098		CALL CROSS(JX1, JY1)		
099		N=IFIRST		
100		IF (LCODE .EO. 1) GO TO 108 IF (KODE) 202,14,109		
101	202	ACCEPI" X2=",JX2," Y2=",JY2		
102	LUC	JX2=JX2/IFACT		
103				

104		JY2=JY2/IFACT
105		GO TO 198
106	109	CALL DIGIT(JX2, JY2, ICODE)
107		GO TO 198
108	14	CALL CURS(I, JX2, JY2) JGET POINT 2
109		CALL CHARO(159)
110		IF(I.NE.178) GOTO 14

111	108	IF(IABS(JX2-JX1).GT.IACC) GOTO 15	
112		IF(IABS(JY2-JY1).GT.IACC) GOTO 15	
113		1F(KODE)202,14,109	
	15	IF(N+LE+1) GOTO 25	
114	15	DO 16 NN=1.N	
115		IF (NN .EQ. IFIRST) GOTO 16	
116			
117		IF(IABS(IX(NN)-JX2).GI.IACC) GOTO 16	
118		IF(IABS(IY(NN)-JY2).GT.IACC) GOTO 16	
119		ISEC=NN	
150		6010 17	
121	16	CONTINUE	
155		GOTO 18	
123	17	JX2=IX(ISEC)	
124		JY2=1Y(ISEC)	
125		CALL CHARO(135)	
126		GOTO 28	
127	18	1F(L.E0.0) GOTO 25	
128		CALL LINEX(JX2, JY2, IXS, IYS, NHIT, LL)	
129		1F(NHIT.EQ.1) GO TO 26	
130	25	ISEC=N+1	
131		GOTO 27	
132	26	JX2=1XS	
133	2.0	JY2=IYS	
134		ISEC=N+1	
135		L=L+1	
136		11(L)=1SEC	
137		12(L)=12(LL)	
138		I2(LL)=ISEC	
139		CALL CHARO(135)	
140	27	IX(ISEC)=JX2	
141		IY(ISEC)=JY2	
142		CALL CROSS(JX2, JY2)	
143		N=ISEC	
144	28	JXD=JX2-JX1	
145		JYD=JY2-JY1	
146		IF(IABS(JYD).GT.IABS(JXD)) GOTO 60	
147		IS%Y=0	
148		IF(JX2.GT.JX1) GOTO 29	
149		6010 49	
150	60	ISWY=1	
151		IE (745+01+741) 0010 54	
152	49	JXL=JX2	
153		JXR=JX1	
154		JYL=JY2	
155		JYR=JY1	
156		IPL=ISEC	
157		IPR=IFIRST	
158		GOTO 30	
159	29	JXL=JX1	
160		JXR=JX2	
161		JYL=JY1	

162 JYR=JY2 163 IPL=IFIRST 164 IPR=ISEC 165 30 IF(ISkY.E0.0)GOTO 61

H=FLOAT (JXR-JXL) /FLOAT (JYR-JYL) 166 NXTOT=0 167 DO 62 NY=1.N 168 IF(IY(NY).GT.JYR.OR.IY(NY).LT.JYL)GO TO 62 169 IF(NY.EQ.IPL.OR.NY.EQ.IPR) GOTO 62 170 IXX=IFIX(H*FLOAT(IY(NY)-JYL))+JXL 171 IF(IABS(IXX-IX(NY)).GT.IACC) GOTO 62 172 173 NXTOT=NXTOT+1 LIST(NXTOT) =NY 174 175 62 CONTINUE 176 GOTO 63 H=FLOAT(JYR-JYL)/FLOAT(JXR-JXL) 61 177 NXTOT=0 D0 31 NX=1.N 178 179 IF(IX(NX).GT.JXR.OR.IX(NX).LT.JXL) GOTO 31 180 IF(NX.EQ.IPL.OR.NX.EQ.IPR) GOTO 31 181 IYY=IFIX(H*FLOAT(IX(NX)-JXL))+JYL 182 IF(IABS(IYY-IY(NX)).GT.IACC) GOTO 31 183 NXTOT=NXTOT+1 LIST(NXTOT)=NX 184 LIST(NXTOT)=NX CONTINUE KOUNT=Ø 185 186 31 187 63 188 C IF(NXTOT-1)50,53,33 IND=0 189 190 33 191 C--ORDER POINT LIST IN INCREASING X (OR Y)--DO 32 NXX=2,NXTOT 192 NX1=LIST(NXX-1) NX2=LIST(NXX) 193 194 NX2=LIST(NXX) IF(ISWY.EQ.1) GOTO 47 195 IF(IX(NX2).GE.IX(NX1)) GOTO 32 196 GOTO 48 197 IF(IY(NX2).GE.IY(NX1)) GOTO 32 47 198 199 48 LIST(NXX-1)=NX2 200 LIST(NXX)=NX1 IND=1 201 IF(IND.EQ.1) GOTO 33 32 CONTINUE 202 203 53 IL=IPL 204 IR=LIST(1) GOTO 51 II=IPI 205 206 50 IL=IPL 207 IR=IPR 208 KOUNT=KOUNT+1 NINT=Ø 209 51 210 LOLD=L DO 35 LK=1,LOLD 211 212 213 C--BEGIN LINE SEARCH FOR THIS SEGMENT--

214	IF 1=11(LK)
215	IF2=12(LK)
216	IF(IF1.EQ.IL.AND.IF2.EQ.IR) GOTO 34
217	IF(IF1.EQ.IR.AND.IF2.EQ.IL) GOTO 34
218	IF(IF1.EQ.IL.OR.IF1.EQ.IR.OR.IF2.EQ.IL.OR.IF2.EQ.IR)GOTO 35
219	CALL OVLAP(IX(IL), IX(IR), IX(IF1), IX(IF2), IX5, IX6, NS1)
220	IF(NS1.EQ.0) GOTO 35

551		CALL OVLAP(IY(IL), IY(IR), IY(IF1), IY(I	F2),	112,11	(6,NS2)
555		IF(NS2.EQ.0) GOTO 35			
223		CALL INSEC(IX(IL), IX(IR), IY(IL), IY(IR) . IX	(IF1),	IX(1F2),
224	*	IY(IF1), IY(IF2), IX5, IX6, IY5, I	Y6,1	NX, INY	.NS3)
225		IF(NS3.E0.0) GOTO 35			
226	CA CR	DSSING HAS BEEN FOUND			
227		N=N+1			
228		IX(N)=INX			
.559		IY(N)=INY			
	CCREA	TE NEW LINE			
231		L=L+1			
232		12(LK)=N			
233		I1(L)=N			
234		12(L)=IF2			
235	CTOTA	CROSSING POINTS INCREMENTED			
236		NINT=NINT+1			
237		LISTC(NINT)=N			
238	35	CONTINUE			
239	00	IF(NINT-1) 41,38,37			
240	37	NIT=Ø			
241	51	DO 36 NN=2, NINT			
242		L1=LISTC(NN-1)			
243		L2=LISTC(NN)			
244		IF(ISWY.EQ.1) GOTO 46			
245		IF(IX(L2).GE.IX(L1)) GOTO 36			
246		GOTO 45			
247	46	IF(IY(L2).GE.IY(L1)) GOTO 36			
248	45	LISTC(NN-1)=L2			
249	45	LISTC(NN)=L1			
250		NIT=1			
251	36	CONTINUE			
252	30	IF(NIT.EQ.1) GOTO 37			
253	38	ILEFT=IL			
254	50	NUT=1			
255	39	L=L+1 (MMA] = 3.00 [
256		II(L)=ILEFT			
257		12(L)=LISTC(NUT)			
258		CALL PLOTS(0, IX(ILEFT), IY(ILEFT))			
259		CALL PLOTS(1, IX(I2(L)), IY(I2(L)))			
260		CALL CROSS(IX(I2(L)),IY(I2(L)))			
261		ILEFT=LISTC(NUT)			
262		IF (NUT.GE.NINT) GOTO 40			
263		NUT=NUT+1			
264		GOTO 39			
265	CLAST	LINE FOR THIS SEGMENT			
266	40	L=L+1			
267		II(L)=ILEFT			
268		12(L)=IR			
040		CALL PLOTS (A. TY/ILEET), TY/ILEETSS			

```
269CALL PLOTS(0,IX(ILEFT),IY(ILEFT))270CALL PLOTS(1,IX(IR),IY(IR))271GOTO 34272 C--NO CROSSINGS ON THIS SEGMENT (JUST ONE LINE TO CREATE)--27341274I1(L)=IL275I2(L)=IR
```

276		CALL PLOTS(0, IX(IL), IY(IL))
277		CALL PLOTS(1, IX(IR), IY(IR))
278	34	IF (KOUNT-NXTOT) 56, 52, 54
279	56	IL=LIST(KOUNT)
280		IR=LIST(KOUNT+1)
281		GOTO 51
282	52	IL=LIST(KOUNT)
283		IR=IPR
284		GOTO 51
285	54	IF(KODE)203,1,100
286	203	CALL CHARO (159)
287		CALL CHARI (MCODE)
288		IF (MCODE . EQ. 197) GO TO 1 J"E" TO ESCAPE NUM. INPUT
289		IF (MCODE.EQ.141) GO TO 201 3 "CR" FOR NEW X1.Y1
290		IF (MCODE .NE. 204) GO TO 301 J"L" TO REDRAW LINES
291		CALL CHARO(155)
292		CALL CHARO(140)
293		DO 302 NL=1.L JREPLOT ARRAY OF LINES
294		IAA=I1(NL)
295		IBB=12(NL)
296		CALL PLOTS(0, IX(IAA), IY(IAA))
297	302	CALL PLOTS(1, IX(IBB), IY(IBB))
298	002	CALL CHARO(159)
299		GO TO 203
300	301	IF (MCODE .NE . 175) GO TO 205 J"/" TO REPEAT POINT
301	501	JX1=JX2
302		JY1=JY2
303		GO TO 103
304	205	TYPE" ?"
305	205	GO TO 203
306	72	CALL HARD
307	• •	GO TO 1
308	73	CALL ERASE(JX1, JY1)
309		GOTO 1
310	74	CALL CHARO(155)
311		CALL CHARO(140)
312		GO TO 75
313	76	CALL CHARI(IN)
314		IF(IN.NE.178) GOTO 1
315		CALL CHARO(155)
316		CALL CHARO(140)
317		LISICIJEN
318		LIST(2)=L
319		LIST(3)=IACC
320		CALL OVLAY(2,11)
321		GO TO 1 ACCEPT " NEW SCALE FACTOR ? " , IFACT
322	210	HOGELT HER DONEL THOTOM T
323		GO TO 1
324		END ; THANK GOODNESS!!!

	CR001	INE TO DETECT IF LINE IS NEAR POINT	
003		COMMON 11(768),12(768),LIST(32),	
004	*	LISTC(128), IX(512), IY(512)	
005		COMMON/HANDY/N,L, IACC	
006		DO 5 LL=1,L	
807		IP1=I1(LL)	
800		IP2=I2(LL)	
309		IX1=IX(IP1)	
010		IY1=IY(IP1)	
011		IX2=IX(IP2)	
312		IY2=IY(IP2)	
013		IYD=IY2-IY1	
014		IXD=IX2-IX1	
015		IF(IABS(IYD).GT.IABS(IXD)) GOTO 6	
316		IF(IX2.GT.IX1) GOTO 7	
317		IF(IXH.LT.IX2.OR.IXH.GT.IX1) GOTO 5	
318	9	H=FLOAT(IYD)/FLOAT(IXD)	
019		IYG=IFIX(H*FLOAT(IXH-IX1)+0.5)+IY1	
950		IF(IABS(IYG-IYH).GT.IACC) GOTO 5	
221		IYR=IYG	
925		IXR=IXH	
223		GOTO 8	
024	7	IF(IXH.LT.IX1.OR.IXH.GT.IX2) GOTO 5	
225		GOTO 9	
326	6	IF(IY2.GT.IY1) GOTO 10	
027		IF(IYH.LT.IY2.OR.IYH.GT.IY1) GOTO 5	
828	11	H=FLOAT(IXD)/FLOAT(IYD)	
829		IXG=IFIX(H*FLOAT(IYH-IY1)+0.5)+IX1	
030		IF(IABS(IXG-IXH).GT.IACC) GOTO 5	
331		IXR=IXG	
32		IYR=IYH	
233		GOTO 8	
334	10	IF(IYH.LT.IY1.OR.IYH.GT.IY2) GOTO 5	
335		GOTO 11	
036	5	CONTINUE	
337		NHIT=0	
338		RETURN	
039	8	NHIT=1	
040		LINE=LL	
341		RETURN	
142		END	
		The state of the second s	

1 66		SUBROUTINE ERASE(IXH, IYH)	
302	CTO	ERASE ONE LINE & RE-DRAW SYSTEM	
03		COMMON I1(768), 12(768), LIST(32),	
304	*	LISTC(128), IX(512), IY(512)	
205		COMMON/HANDY/N, L, IACC	
396		CALL LINEX(IXH, IYH, IXR, IYR, NHIT, LINE)	
307		IF (NHIT.EQ.Ø) RETURN	
	CERA:	SE SCREEN	
09		CALL CHARO(155)	
10		CALL CHARO(140)	
11	CCUT	OUT LL; SHUFFLE DOWN REST	
12		LL=LINE	
13		IF(LL.EQ.L) GOTO 2	
14			
15		DO 1 LK=LL,L1	
17	1	I1(LK)=I1(LK+1)	
18	21	I2(LK) = I2(LK+1) L=L-1	
19	-	DO 3 LX=1.L	
80		IA=I1(LX)	
21		IB=I2(LX)	
55		CALL PLOTS(0, IX(IA), IY(IA))	
23	3	CALL PLOTS(1, IX(IB), IY(IB))	
24		CALL CHARO(159)	
25		RETURN	
26		END	

001		SUBROUTINE INSEC(IX1, IX2, IY1, IY2, IX3, IX4, IY3, IY4,
002		<pre>* IX5,IX6,IY5,IY6,IX,IY,NSUC)</pre>
003		ID1=IX2-IX1
004		ID2=IY2-IY1
005		ID3=IX4-IX3
006		ID4=IY4-IY3
007		IF(ID1.EQ.0) GO TO 1
008		IF(ID2.E0.0) GO TO 2
009		IF(IABS(ID2).E0.IABS(ID1)) GO TO 3
010		IF(IABS(ID1).GT.IABS(ID2)) GO TO 4
Ø11	10	IF(IABS(ID3).GT.IABS(ID4)) GO TO 14
012		H1=FLOAT(ID1)/FLOAT(ID2)
Ø13		IX1L=IFIX(H1*FLOAT(IY5-IY1))+IX1
014		IX1R=IFIX(H1*FLOAT(IY6-IY1))+IX1
015		G2=FLOAT(ID3)/FLOAT(ID4)
016		IX2L=IFIX(G2*FLOAT(IY5-IY3))+IX3
017		IX2R=IFIX(G2*FLOAT(IY6-IY3))+IX3
018		IXDL=IX2L-IX1L
019		IXDR=IX2R-IX1R
020		IF(ISIGN(1,IXDL).EQ.ISIGN(1,IXDR)) GO TO 99
021		R=FLOAT(IABS(IXDL))/FLOAT(IABS(IXDR-IXDL))
022		IY=IY5+IFIX(R*FLOAT(IY6-IY5))
023		IX=IFIX(H1*FLOAT(IY-IY1))+IX1
024		NSUC=1
025		RETURN
026	14	H1=FLOAT(ID1)/FLOAT(ID2)
027		IF(ID4.EQ.0) GO TO 15
028		G1=FLOAT(ID4)/FLOAT(ID3)
029		GH=G1*H1
030		IY=(G1*FLOAT(IX1-IX3)-GH*FLOAT(IY1)+FLOAT(IY3))/(1.Ø-GH)
031	17	IX=IFIX(H1*FLOAT(IY-IY1))+IX1
032	16	IF((IX.GT.IX6).OR.(IX.LT.IX5)) GO TO 99
033		IF((IY.GT.IY6).OR.(IY.LT.IY5)) GO TO 99
034		NSUC=1
035		RETURN
036	15	IY=IY3
037		GO TO 17
038	1	IF(ID4.NE.0) GO TO 10
039	-	IX=IX1
040		IY=IY3
041		NSUC=1
042		RETURN
043	2	IF(ID3.NE.0) GO TO 4
044		1X=1X3
045		
046		NSUC=1
040		RETURN
047	2	IF(IABS(ID4).EQ.IABS(ID3)) GO TO 99
127.01 (22.3	100	
049	4	IF(IABS(ID3).GT.IABS(ID4)) GO TO 12

050 H2=FLOAT(ID2)/FLOAT(ID1) 051 IF(ID3.EG.0) GO TO 18 052 G2=FLOAT(ID3)/FLOAT(ID4) 053 GH=G2*H2 054 IX=(G2*FLOAT(IY1-IY3)-GH*FLOAT(IX1)+FLOAT(IX3))/(1.0-GH) 055 19 IY=IFIX(H2*FLOAT(IX-IX1))+IY1

```
056
          GO TO 16
057
       18 IX=IX3
          GO TO 19
058
       12 H2=FLOAT(ID2)/FLOAT(ID1)
059
          IYIL=IFIX(H2*FLOAT(IX5-IX1))+IY1
060
          IY1R=IFIX(H2*FLOAT(IX6-IX1))+IY1
061
062
          G1=FLOAT(ID4)/FLOAT(ID3)
          IY2L=IFIX(G1*FLOAT(IX5-IX3))+IY3
063
          IY2R=IFIX(G1*FLOAT(IX6-IX3))+IY3
064
065
          IYDL=IY2L-IY1L
          IYDR=IY2R-IY1R
066
          IF(ISIGN(1,IYDR).EQ.ISIGN(1,IYDL)) GO TO 99
067
          R=FLOAT(IABS(IYDL))/FLOAT(IABS(IYDR-IYDL))
068
069
          IX=IX5+IFIX(R*FLOAT(IX6-IX5))
070
          IY=IFIX(H2*FLOAT(IX-IX1))+IY1
          NSUC=1
071
072
          RETURN
073
       99 NSUC=0
074
          RETURN
          END
075
```

001		SUBROUTINE HARD	
002	CROUT	TINE TO MAKE A HARD COPY OF DISPLAY	
003		COMMON 11(768), 12(768), LIST(32),	
004	*	LISTC(128), IX(512), IY(512)	
005		COMMON/HANDY/N,L, IACC	
006		CALL COPY (ISWIT) JSWITCH OFF=4631	
007			
		IF(ISWIT .EQ. 0) GO TO 5	
008		DO 1 K=1,L	
009		IP1=I1(K)	
010		IP2=12(K)	
011		MX=4*IX(IP1)-2047	
015		MY=4*IY(IP1)-2047	
013		CALL PLOT(MX,MY,3)	
014		MX=4*IX(IP2)-2047	
015		MY=4*IY(IP2)-2047	
016	1	CALL PLOT (MX, MY, 2)	
017		DO 2 J=1,N	
018		MX=4*IX(J)-2017	
019		MY=4*IY(J)-2017	
020	2	CALL INUM (MX, MY, J, 4)	
021		CALL PLOT (-2047, -2047, 3)	
022	5	CONTINUE	
023	100	RETURN	
024		END	
004		LID	

NOTE: PLOT IS THE SUBROUTINE DESCRIBED BY CUNDALL (1974) FOR PLOTTING THE LINES OR BLOCKS ON AN X-Y RECORDER

001	SUBROUTINE CROSS(IX, 1Y)
002	CALL PLOTS(0, IX+10, IY)
003	CALL PLOTS(1, IX-10, IY)
004	CALL PLOTS(0, IX, IY+10)
005	CALL PLOTS(1, IX, IY-10)
006	CALL CHARO(159)
007	RETURN
008	END

C-14

161						C-15
		.TITL	TEK			
		. ENT	CHARO, CH	ARI .CL	RS, PLOTS	As bacon dappa
		.EXTD	.FRET		The second	ALCONDING TO A
		.NREL				
177611		N=-167				
177612		N1=N+1				
177613		N2=N1+1				
00000.0000055		2				
00001 '0060025	CHARO:	JSR	e.CPYL			
00002 . 060277	NA, CELES	INTDS	THEFT			
00003'027611		LDA	1, eN, 3			
00004 0 44407		STA	1.TWIT			
00005'004451		JSR	CHOUT			
00006'000013'		TWIT	10191			
00007 060177		INTEN				
00010.0060015		JSR	e.FRET			
00011'000000	TWET:	Ø	A COLORED			
00012.000000	TWOT:	ø				
00013'000000	TWIT:	ø				
00014'000000	SV3:	õ				
00015.000002		2				
00016'0060025	CHARI:	JSR	0.CPYL			
00017'054775		STA	3,5V3			
00020'060277		INTDS				
00021 004426		JSR	CHIN			
00022.000013.		TWIT				
00023'024770		LDA	1,TWIT			
00024'034770		LDA	3, SV3			
00025'047611		STA	1, eN, 3			
00026'060177		INTEN				
00027 0060015		JSR	e.FRET			
00030 000004		4				
00031 0060025	PLOTS:	JSR	e.CPYL			
00032'060277		INTDS				
00033'027611		LDA	1.0N.3			
00034'044757		STA	1.TWIT			
00035'027612		LDA	1, eN1,3			
COOCH CLAPTER		C T A	a second s			

JSR e.FRET 0, CCACØ ;SAVE ACØ STA ISKP IF CHAR READY TTI SKPDN JMP . - 1 FREAD CHAR Ø,TTI DIAS STORE CHAR STA 0,00,3 0,CCACØ ;RESTORE ACO LDA

1.TWET

TPLOT

1,3

1,3

1,0N2,3 1,TWOT

00056'040407 00057 063511 00060'000777 00061 023400 00062'061111 00063'020402 00064'001401

00036 044753

00037 027613

00040 044752

00041 004425

00042 000013

00043'000011'

00044'000012'

00045'060177

00046'0060015

00047 040416

00050'063610 00051 000777

00052 060510

00053'043400

00054'020411

00055'001401

CHOUT: STA SKPBE JMP LDA DOAS LDA JMP

JMP

STA

LDA

STA JSR

TWIT

TWET

TWOT

CHIN:

INTEN

0, CCACO ISAVE ACO JSKIP IF NOT BUSY TTO • - 1 JGET CHARACTER 0,00,3 JSHIP CHARACTER Ø,TTO 0,CCACØ JRESTORE ACØ

IRETURN

C-16 TEMP FOR ACØ SAVE ACØ GET X

STA TPLOT: Ø, TPTACO; SAVE ACØ 00066'040526 JGET X LDA 0,01,3 00067 023401 Ø, TPTX STA 00070'040526 JGET Y 00071 023402 LDA 0,02,3 STA Ø, TPTY 00072'040525 JGET MODE 00073'023400 LDA 0,00,3 00074'040524 STA Ø, TPMOD 3, TPTADD; SAVE CALL ADDRESS 00075'054520 STA 0.0. SNR ISKP IF NEG Ø 00076'101015 MOV# INITIALIZE AND DARK VECTOR 1= 0 00077 000405 JMP TPTDV 0,0,SNC ;SKIP IF < 0 00100'101113 MOVL# INORMAL BRIGHT VECTOR 00101 000405 JMP TPTNRM **echous** SET TO ALPHA 00102'006511 JSR 00103'000202' US JDARK VECTOR **eCHOUZ** 00104'006507 TPTDV: JSR GS 00105'000201' JGET Y 00106'020511 TPTNRM: LDA Ø, TPTY 0,0,SEC 1SKP IF + MOVL# 00107'101112 SUB IMAKE Ø 00110102400 0.0 JUPPER Y BOUND 001111 034477 LDA 3,D780 3,0, SNC ; SKP IF ON SCREEN 00112'162513 SUBL# JSET TO EDGE MOV 00113'161000 3,0 SAVE GOOD Y STA Ø, TPTY 00114'040503 00115'101120 MOVEL 0,0 JUSE UPPER 5 BITS 00116'101120 MOVEL 0.0 00117 101120 MOVEL 0,0 JAND SWAP HALVES 00120.101300 MOVS 0,0 JHI Y TAG 00121 034463 LDA 3,8040 JPUT IN CHAR 00122'163000 ADD 3,0 00123'040476 Ø, TPTTMP; USE A TEMP STA JSR **ECHOUE** SHIP HI Y 5 00124 006467 00125'000221' TPTTMP Ø, TPTY JGET Y 00126'020471 LDA 00127 034453 LDA 3,BØ37 ; MASK 00130'163400 ILEAVE LOW Y 5 AND 3,0 ILOW Y TAG LDA 00131 034455 3,B140 SET IN CHAR 00132'163000 ADD 3,0 STA Ø, TPTTMP 00133'040466 JSR **e**CHOUZ SHIP LOW Y 00134'006457 00135'000221' TPTTMP 00136'020460 LDA Ø, TPTX JGET X VALUE 00137'101112 MOVL# 0,0,SZC 00140'102400 SUB 0.0 00141 034450 LDA 3,D1023 00142'162513 SUBL# 3,0,SNC 00143'161000 MOV 3,0 00144'040452 Ø, TPTX STA 00145'101120 JAND DO LIKE Y MOVEL 0,0 MOVEL 00146'101120 0,0 00147 101120 MOVEL 0,0 00150'101300 MOVS 0,0 JHI X 5 JHI X TAG 00151 034433 LDA 3,8040 00152'163000 JADD IN TAG ADD 3,0 00153'040446 STA Ø, TPTTMP **eCHOUZ** SHIP HI X 5 JSR 00154'006437 00155'000221' TPTTMP 00156'020440 LDA Ø, TPTX JGET X LDA 3,8037 JGOODIE MASK 00157 034423 00160'163400 ILEAVE LOW X 5 AND 3,0

00065'000000

CCACØ:

Ø

00161 034424		LDA	3,B100	JLOW X TAG
00162 163000		ADD	3,0	JPUT IN TAG
00163'040436		STA	Ø, TPTTM	P
00164'006427		JSR	eCHOUZ	
00165'000221'		TPTTMP		
00166'020432		LDA	Ø, TPMOD	
00167'101113		MOVL#	0,0,SNC	
00170'000404		JMP	TPTEXT	
00171'102400		SUB	0.0	
00172'040426		STA	Ø, TPMOD	
00173 000713		JMP	TPTNRM	
00174'020420	TPTEXT:	LDA		ØJRESTORE ACØ
00175'034420		LDA		DICALL ADDRESS
00176'001403		JMP	3,3	JEXIT
00177 000032	SUBOQ:	032	0,0	
00200'000033	ESC:	033		
	GS:	035		
00201'000035				
00202'000037	US:	037		
00203'000020	B020:	020		
000202'	B037=US	0.10		
00204'000040	BØ40:	040		
00205 000100	B100:	100		
00206'000140	B140:	140		
00207 000003	D003:	003		
00210'001414	D780:	1414		
00211 001777	D1023:	1777		
00212'000047'	CHINP:	CHIN		
00213'000056'		CHOUT		
00214'000000		0		
00215'000000	TPTADD:			
00510.000000	TPTX:	Ø		
00217 000000	TPTY:	0		
00550.000000	TPMOD:	Ø		
00221.000000	TPTTMP:	Ø	-	
00222*040772	CURSIS:	STA		ØJSAVE ACØ
00223'054772		STA		D; SAVE CALL ADDRESS
00224'006767		JSR	echou _f	SET TO ALPHA
00225.000505.		US		
00226'006765		JSR	e CHOUZ	; TURN ON CURSER
00227'000200'		ESC	an anna	
00230'006763		JSR	e CHOUZ	
00231'000177'		SUBOO		
00232'006760		JSR	eCHINP	JGET CHAR
00233'000216'		TPTX		
00234'020753		LDA	0,D003	JGET LOOP COUNTER
00235'040764		STA	Ø,TPTTM	
00236'020760		LDA	Ø,TPTX	JGET CHAR
00237 000421		JMP	CURPS	STORE CHAR
00240'006752	CURLP:	JSR	e CHINP	JGET HI COORD
00241'000216'		TPTX		
00242'006750		JSR	eCHINP	JGET LOW COORD
00243'000217'		TPTY		

00244'034736 00245'020752 00246'163400 00247'040750 00250'020746 00251'163400 00252'101300 00253'101220 LDA LDA AND STA LDA AND MOVS MOVZR 3,8037 ;MASK 0,TPTY ;LOW COORD 3,0 ;MASK OFF GARBAGE 0,TPTY ;SAVE FOR LATER 0,TPTX ;HI COORD 3,0 ;MASK OFF 0,0 ;SWAP 0,0

1.5							1
	00254'101220		MOVER	0.0			
	00255'101220		MOVER	0.0			
	00256'034741		LDA	3, TPTY	JLOW COO	RD	
	00257'163000		ADD	3,0		LON COORD	
	00260 034735	CURPS:	LDA		CALL AD	And the second se	
	00261 043400	00111 5.	STA			STRATIC CLARKE	
	00262'175400			0,00,3		the second se	
	00263'054732		INC	3,3	JADJUST	A COMPANY AND A CO	
	Statement in the statement		STA			DATED ADD	
	00264'014735		DSZ	TPTTMP	CHECK F	- 19	
	00265'000753		JMP	CURLP	;LOOP IF		
	00266'020726		LDA		JI RESTORE	ACØ	
	00267'001400		JMP	0,3	1 RETURN		
	00270 000004		4				
	00271 0060025	CURS:	JSR	e.CPYL			
	00272'060277		INTDS				
	00273 054416		STA	3, SX3			
	00274'004726		JSR	CURSIS			
	00275'000312'		A1	No. C. C. C. C. C.			
	00276'000313'		A2				
	00277'000314'		A3				
	00300 034411		LDA	3, 5X3			
	00301'024411		LDA	Ser Strates			
	00302'047611			1,A1			
			STA	1, eN, 3			
	00303 024410		LDA	1,A2			
	00304'047612		STA	1, eN1, 3			
	00305'024407		LDA	1,A3			
	00306'047613		STA	1, eN2, 3			
	00307 060177		INTEN				
	00310'0060015		JSR	0.FRET			
	00311'000000	SX3:	Ø				
	00312'000000	A1:	0				
	00313'000000	A2:	Ø				
	00314'000000	A3:	Ø				
			.END				

-

		.TITL	TAPE	C-19
		.ENT	TAPE, O	VLAY
		.EXTD	. CPYL	FRET
		.NREL		
177611		N=-167		
00000.000000	NUB:	0		
800001 '000002	TWO:	2		
00002.000003	THREE:	3		
00003'000000'	FIRST:	NUB		
00004'000322'	LAST:	CS		
00005 000003	200928 222	3		
	JTHIS R	OUTINE RI	EADS THE	APPROPRIATE OVERLAY
	JFROM T	APE. IT	STARTS	BY FIRST TRANSFERING
	JITSELF	TO A SAL	FE PLACE	IN HIGH CORE.
00006'0060015	OVLAY:	JSR	0.CPYL	
00007 060277		INTDS		
00010'020476		LDA	Ø, DRIVE	
00011 062074		DOB	Ø.LINC	
00012'054473		STA	3. SAVE	
00013'023611		LDA	Ø. 8N. 3	
00014'040764		STA	Ø,NUB	JOVERLAY NUMBER
00015'035612		LDA	3,N+1,3	JADDR OF LOWEST ARRAY
00016'030765		LDA	2,FIRST	
00017'020765		LDA	Ø.LAST	
00020'142400		SUB	2,0	J=NUMBER OF WORDS TO BE MOVED
00021'101400		INC	0,0	
00022'116400		SUB	0.3	JADDR TO MOVE TAPE ROUTINE TO
00023'100400		NEG	0.0	
00024'025000	ROUND:	LDA	1,0,2	
00025'045400		STA	1,0,3	
00026'101405		INC	0.0. SNR	
00027 000404		JMP	OUT	
00030'151400		INC	2,2	
00031'175400		INC	3,3	
00032'000772		JMP	ROUND	
00033'156400	OUT:	SUB	2,3	J=DISTANCE MOVED
00034'030403	100 March 100	LDA	2,SHIFT	State of the state
00035'157000		ADD	2,3	
00036'001400		JMP	0,3	J GO TO HI-CORE COPY
00037'000040'	SHIFT:	.+1		CALLS AND
00040'020740		LDA	Ø,NUB	
00041'126520		SUBEL	1+1	
00042'122415		SUB#	1,0, SNR	
00043'000407		JMP	Al	SOVERLAY 1
00044'024735		LDA	1,TWO	REAL ADJ AND THE REAL OF CARES
00045'122415		SUB#	1,0, SNR	
00046'000407		JMP	A2	SOVERLAY 2
00047 020434		LDA	Ø.BLK3	SOVERLAY 3
00050'024434		LDA	I NBLK3	LITHOLDS & ROT WORK
00051 000406		JMP	CAT	
00052'020425	A1:	LDA	Ø,BLKI	
00053'024425	2425	LDA	I.NBLKI	
00054'000403		JMP	CAT	
00055'020424	A2:	LDA	Ø,BLK2	
00055 020424				

00055'020424 00056'024424 00057'152400 00060'034415 00061'054452 00062'004411 00063'125005 LDA SUB LDA STA JSR MOV

CAT:

0,BLK2 1,NBLK2 2,2 3,SUBST 3,RETRN NIXON 1,1,SNR

1024-03199-109 5024-05104-109 611-05105-05109 605-05105-05109 615-0511-10109 615-0511-10109

-	00064.000377		JMP	377	FORT	RAN ST	ART ADI	RESS
	00065'063077		HALT		ILINC			States and the second
	00066'020420		LDA	A.DRIVE	C.C.C.C.C.C.C.C.C.C.C.C.C.C.C.C.C.C.C.	and the second s		CONTINUE)
	00067 062074		DOB	ØLINC		. Crian	II NEDD	
	00070'000750		JMP	SHIFT+1				
		Noco	and the second se	SHIFITI				
	00071'060177	NOGO:	INTEN	A FRET				
	00072'0060025	NT YON .	JSR	e.FRET				
	00073'054412	NIXON:	STA	3, SAVE				
	00074'000445		JMP	RLINC				
	00075'002752	SUBST:	JMP	ESAVE-RE	EIRN I	1 20B2	TITUIE	CONTENTS FOR
	00076'000000	ORIG:	Ø					
	00077 000350	BLK1:	350					
	00100'000055	NBLK1:	55					
	00101 000450	BLK2:	450					
	00102'000037	NBLK2:	37					
	00103.000210	BLK3:	510					
	00104'000037	NBLK3:	37					
	00105.000000	SAVE:	Ø					ARTESSALLESSA
	00106'000001	DRIVE:	1					
	00107'000006		6					
		and the second sec		NABLES A			A DESCRIPTION OF THE OWNER OF THE	
		ITO WRI	TE BLOCK	S OF CORE	ONTO	TAPE.		
		1						
	00110'0060015	TAPE:	JSR	e.CPYL				
	00111'060277		INTDS					
	00112'102400		SUB	0,0				
	00113'062074		DOB	Ø,LINC				
	00114'054771		STA	3, SAVE				
	00115'023612		LDA	Ø, EN+1,3	3			
	00116'027613		LDA	1, eN+2,3				
	00117'031614		LDA	2,N+3,3				
	00120'037611		LDA	3, eN, 3				
	00121'175005		MOV	3,3, SNR				
	00122'000415		JMP	CLINC				
	00123175112		MOVL#	3,3,SEC				
	00124'000404		JMP	NEGA				
	00125'175234	DOG:	MOVER#	3,3,52R				
	00126'000415		JMP		JMUST			
	00127'000412		JMP		IMUST	BE 1		
	00130'174400	NEGA:	NEG	3,3				
	00131'150000		COM	2,2				
	00132'000773		JMP	DOG				
	00133'034752	RETRN:	LDA	3, SAVE				
	00134 047615		STA	1, eN+4,3	3			
	00135'060177		INTEN					
	00136'0060025		JSR	e.FRET				
		INOW FO		HTLY MODI	FIED V	ERSIO	N OF TH	E
				TAPE UTIL				ARAMAN'S LAND
	001371152400	CLINC:	SUB	2,2				
	00140'000415	outivo.	JMP	CHKE				
	00141 034426	RLINC:		3,D2R				
		ALINC:	LDA					
	00142'000414		JMP	READZ				

DDIAE DDDAIA	
00143'034422	WLINC:
00144'054507	
00145'044500	
00146'050416	
00147'004422	
00150'024475	RAW:
00151'122400	
00152'030412	

JMP LDA STA STA JSR LDA SUB LDA READZ 3,D1W 3,D1XX 1,D2XX 2,SAC2 D0 1,D2XX 1,0 2,SAC2

-					
	00153'151113		MOVL#	2,2,SNC	
	00154'150000		COM	2,2	
	00155'034472	CHKZ:	LDA	3,020	
	00156'054467	READZ:	STA	3.D2XX	
	00157 034407		LDA	3,D1RC	
	00160'054473		STA	3,D1XX	
				and the first second	
	00161'004410		JSR	DO	
	00162'060274	EXIT:	NIOC	LINC	
	00163'000750		JMP	RETRN	
	00164'000000	SAC2:	0		
	00165 021000	D1W:	LDA	0,0,2	
	00166'000750	DIRC:	JMP	READ-DIXX,1	
	00167'132512	D2R:	SUBL#	1,2,SZC	
	00170'000000		Ø		
	00171 054777	DO:	STA	3, RETU	
	00172'075474	00.	DIB	3.LINC	
	00173'175112		MOVL#	3,3,SEC	
			2010	EA	
	00174'000446		JMP	and the second se	
	00175'151113		MOVL#	2,2, SNC	
	00176'000410		JMP	FINDF	
	00177 150000		COM	2,2	
	00200'176400	FINDR:	SUB	3,3	
	00201 162000		ADC	3,0	
	00202 060374		NIOP	LINC	
	00203 004467		JSR	GETBL	
	00204'101401	FINDN:	INC	0,0,SKP	
	00205 000776		JMP	2	
	00206'060174	FINDE.	NIOS	LINC	
	The second s	FINDF.	JSR	GETBL	
	00207'004463				
	00210'000777		JMP	-1	
	00211 175224		MOVER	3,3,SER	
	00212'000766		JMP	FINDR	
	00213'125005	FOUND:	MOV	1,1, SNR	
	00214'002754		JMP	eretu	
	00215'166000		ADC	3,1	
	00216*040474		STA	Ø,TEMP1	
	00217 044474		STA	1, TEMP2	
	00220 024476		LDA	1,SIZE	
	00221'147000		ADD	2,1	
	00222'000431		JMP	DIXX	
	00223 063674	READ:	SKPDN	LINC	
	00224'000777		JMP	+=1	
	00225'063474		SKPBN	LINC	
	00226'000416		JMP	RDAT	
		DOUV.		ØLINC	
	00227'060474	RCHK:	DIA		
	00230'116405		SUB	0,3,5NR	
	00231 000434		JMP	SCHK	
	00232*024465	E1:	LDA	1.01	
	00233'000403		JMP	•+3	
	00234 034462	E2:	LDA	3,SIZE	
	00235 024463		LDA	1,02	
	00236'020454		LDA	Ø,TEMP1	
	00237 000723		JMP	EXIT	
	00240'024461	E3:	LDA	1,04	
	00241 000721		JMP	EXIT	
	00242'024460	E4:	LDA	1,08	
	00243 000717		JMP	EXIT	
	00244'060474	RDAT:	DIA	0,LINC	
	00245 132512		SUBL#	1,2,SZC	
		D2XX:		And the state of the second	
	00246'041000		STA	0,0,2	

00247 000402	DSC:	JMP	.+2	CT-I LOL SOLDING
00250 061074	WDAT:	DOA	Ø,LINC	C-22
00251 117000	BLOOP:	ADD	0.3	
002521151400		INC	2,2	
00253 021000	D1XX:	LDA	0,0,2	
00254'063074		DOC	Ø.LINC	
00255'063674		SKPDN	LINC	
00256'000777		JMP	1	
00257 063474		SKPBN	LINC	
00260 000770		JMP	WDAT	
00261 075074	WCHK:	DOA	3.LINC	
00262'075474		DIB	3,LINC	
00263'175004		MOV	3,3,SZR	
00264'000756		JMP	E4	
00265'132414	SCHK:	SUB#	1,2, SER	
00266'000746	DUTIT	JMP	E2	
00267 020423	NEXT:	LDA	Ø,TEMP1	
00270'024423		LDA	1. TEMP2	
00271'000713		JMP	FINDN	
00272'054420	GETBL:	STA	3.TEMP1	
00273'034421	02102.	LDA	3.MLIM	
00274'162432		SUBZ#	3,0,SZC	
00275'000405		JMP	WAIT	
00276'034417		LDA	3.PLIM	
00277'162032		ADCZ#	3,0,SEC	
00300'000740		JMP	E3	
00301 074474		DIA	3.LINC	
00302'063474	WAIT:	SKPBN	LINC	
00303'000777		JMP	WAIT	
00304'063774		SKPDZ	LINC	
00305'000774		JMP	WAIT-1	
00306'074474		DIA	3.LINC	
00307'116543		SUBOL	0,3,5NC	
00310'010402		ISZ	TEMP1	
00311'002401		JMP	eTEMP1	
00312'000000	TEMP1:	Ø		
00313'000000	TEMP2:	ø		
00314'177770	MLIM:	177770		
00315'000620	PLIM:	620		
00316'000400	SIZE:	400		
00317 000001	C1:	1		
00320.000005	C2:	2		
00321'000004	C4:	4		
00322'000010	C8:	10		
	Total Contraction	-FND		

		.TITL	COPY		C 22
		.ENT	COPY		C-23
		.EXTD	.CYPL	FRFT	
		.NREL			
177611		N=-167			
00000.000002		2			
00001 0060015	COPY:	JSR	e.CYPL		
00002.054422		STA	3.ACSV		
00003 060477		READS	Ø	ICHECK FOR SWITCH	0
00004'101122		MOVEL	0,0,SEC	JOFF=4631 ON=PLOTT	CP
00005 000414		JMP	PLTR		C.N
00006'020417		LDA	Ø,ESC		
00007 063511		SKPBE	TTO		
00010'000777		JMP	1		
00011'061111		DOAS	Ø,TTO		
00012'020414		LDA	Ø,ETB		
00013 063511		SKPBZ	TTO		
00014 000777		JMP	1		
00015'061111		DOAS	Ø,TTO		
00016'102440		SUBO	0.0		
00017 043611		STA	0, eN, 3	PUT A ZERO SO HAR	D SKIPS
00020.000403		JMP	BACK	and the second second second	
00021 102520	PLTR:	SUBEL	0.0	PUT A ONE TO PLOT	5
00022 043611		STA	Ø. 8N. 3		
00023 0060025	BACK:	JSR	e.FRET		
	3				
00024 000000	ACSV:	0			
00025 000033	ESC:	27.			
00026.000051	ETB:	23.			
	3				
		. END			

		.TITL	OVLAP	
		.ENT	OVLAP	
		•EXTD	.CPYL, .FRET	
		.NREL		
177611		N=-167		
177612		N1=N+1		
177613		N2=N+2		
177614		N3=N+3		
177615		N4=N+4		
177616		N5=N+5		
177617		N6=N+6		
00000.000000	SAVE:	0		
00001 000000	X5:	Ø		
00002.000000	X6:	Ø		
00003.000010		10		
00004'0060015	OVLAP:	JSR	e.CPYL	
00005'054773		STA	3, SAVE	
00006'023611		LDA	0, eN, 3	
00007 027612		LDA	1, eN1, 3	
00010 033613		LDA	2, eN2, 3	
00011 037614		LDA	3, 0N3, 3	
00012'122512		SUBL#	1,0,SZC	
00013'000455		JMP	F1	
00014'172512		SUBL#	3,2,SEC	
00015'000426		JMP	F2	
00016'162513		SUBL#	3,0,SNC	
00017'132512		SUBL#	1,2,SEC	
00020 000533		JMP	NOGO	
00021'112512		SUBL#	0,2,SEC	
00022'000411		JMP	F3	
00023'136512		SUBL#	1,3,SEC	
00024'000404		JMP	F4	
00025'054754		STA	3,X5	
00026'040754		STA	Ø, X6	
00027 000514		JMP	OK	
00030'044751	F4:	STA	1,X5	
00031'040751		STA	Ø,X6	
00032'000511		JMP	OK	
00033'136512	F3:	SUBL#	1,3,SEC	
00034'000404	2.5	JMP	F5	
00035'054744		STA	3, X5	
00036'050744		STA	2, X6	
00037'000504		JMP	OK	
00040'044741	F5:	STA	1, X5	
00041 050741		STA	2, X6	
00042 000501		JMP	OK	
00043'142513	F2:	SUBL#	2,0,SNC	
00044'136512		SUBL#	1,3,SEC	
00045'000506		JMP	NOGO	
00046'116512		SUBL#	0,3,52C	
00047'000411		JMP	F6	
00050'132512		SUBL#	1,2,520	
00051'000404		JMP	F7	
00051 000404		STA	2, 15	

00025 020151	
00053'040727	
00054'000467	
00055'044724	F7:
00056'040724	
00057'000464	
00060'132512	F6:

STA JMP STA STA JMP SUBL#

0,X6 0K 1,X5 0,X6 0K 1,2,SEC

00061 000404		JMP	F8	
00062 050717		STA	2, X5	
00063'054717		STA	3, X6	
00064'000457		JMP	OK	
00065'044714	F8:	STA	1,X5	
00066'054714	1. Carlotte	STA	3, X6	
00067 000454		JMP	OK	
00070'172512	F1:	SUBL#	3,2,SEC	
00071'000426		JMP	F9	
00072'166513		SUBL#	3,1, SNC	
00073'112512		SUBL#	0,2,SZC	
00074'000457		JMP	NOGO	
00075'132512		SUBL#	1,2,520	
00076'000411		JMP	F10	
00077 116512		SUBL#	Ø,3,SEC	
00100'000404		JMP	F11	
00101'054700		STA	3, X5	
00102'044700		STA	1, X6	
00103'000440		JMP	OK	
00104'040675	F11:	STA	Ø, X5	
00105'044675		STA	1, X6	
00106'000435		JMP	OK	
00107'116512	F10:	SUBL#	0,3,SZC	
00110'000404		JMP	F12	
00111'054670		STA	3.X5	
00112'050670		STA	2,X6	
00113'000430		JMP	OK	
00114'040665	F12:	STA	Ø, X5	
00115'050665		STA	2, X6	
00116'000425		JMP	OK	
00117'146513	F9:	SUBL#	2,1, SNC	
001201116512		SUBL#	Ø,3,SEC	
00121'000432		JMP	NOGO	
00122'136512		SUBL#	1,3,SEC	
00123'000411		JMP	F13	
00124'112512		SUBL#	0,2,52C F14	
00125'000404 00126'050653		JMP STA	2, 14	
00127 044653				
00127 044655		STA	1,X6 OK	
00131 040650	F14:	JMP	Ø, X5	
00132'044650	P 14:	STA	1, X6	
00133'000410		JMP	OK	
00134'112512	F13:	SUBL#	0,2,SEC	
00135'000404	r13.	JMP	F15	
00136'050643		STA	2, 15	
00137 054643		STA	3, X6	
00140'000403		JMP	OK	
00141'040640	F15:	STA	Ø, X5	
00142'054640		STA	3,X6	
00143'020636	OK:	LDA	Ø,X5	
00144'024636	uni	LDA	1, X6	
00145'034633		LDA	3. SAVE	
00146'043615		STA	0, EN4,3	
00147'047616		STA	1, eN5,3	
00150'102520		SUBZL	8,0	
00151'043617		STA	0, EN6, 3	
00152'0060025		JSR	8.FRET	
00153'034625	NOGO:	LDA	3. SAVE	
00154'102460		SUBC	0.0	
		and the second s	000000000	

00155'043617	CTA	0 0010	~	
	STA	0. eN6.		
00156'0060025	JSR	e.FRET		
	. END			

		•TITL •ENT	DIGIT	C-27
		•EXTD	·CPYL	RET
	;	Report of		
			BY PAC	DIGITIZER ROUTINE
	0.772 (0.470)-52.01.32			TO ACCOMODATE ANALOG
	3			
	;	.NREL		
177611	N=-167	NNEL		
000041	DVCE=41		INO LONG	SER DEVICE 42
00000'002400	MODE :	2400		
00001 '000004		4		
00002 0060015	DIGIT:	JSR	0.CPYL	
00003'060277		INTDS		
00004'020774		LDA	Ø.MODE	
00005'062041		DOB	Ø.DVCE	
00006'000457		JMP	BACK	
00007 063710	LOOP:	SKPDZ	TTI	
00010'000466		JMP	HIT	
00011'020476		LDA	Ø,CH3	INO LONGER CHANNEL Ø
00012'061041		DOA	Ø, DVCE	
00013'063641		SKPDN	DVCE	
00014'000777		JMP	• - 1	
00015'060441		DIA	Ø, DVCE	
00016'024466		LDA	1.01000	
00017'106513		SUBL#	Ø,1,SNC	
00020'000767		JMP	LOOP	
00021'020464		LDA	Ø,CH1	
00022'061041		DOA	Ø, DVCE	JGET X
00023 063641		SKPDN	DVCE	
00024'000777		JMP	•-1	
00025'060441		DIA	Ø, DVCE	
00026'043611		STA	0, EN, 3	
00027'020457		LDA	Ø.CH2	
00030'061041		DOA	Ø.DVCE	
00031'063641		SKPDN	DVCE	
00032'000777		JMP	•-1	
00033'060441		DIA	Ø, DVCE	
00034'043612		STA	Ø, eN+1, 3	3
00035'102400		SUB	0.0	.7550 505 10055
00036'043613		STA	Ø, eN+2,3	3 JZERO FOR ICODE
00037'020422		LDA	Ø,MAX	POUTINE TO FLASH LAND
00040'024422		LDA		ROUTINE TO FLASH LAMP
00041'063634		SKPDN	34	JWHEN ACKNOWLEDGING DATA
00042'000777		JMP	1	JINTO BLOCKS PROGRAM
00043'066034		DOB	1,34	
00044'061034		DOA	0,34	
00045'020416		LDA	Ø, DEL	
00046'040416	DELAVA	STA	Ø, COUNT	
00047'060000 00050'060000	DELAY:	NIO	0	
00000 000000		N I U		

00051'014413 00052'000775 00053'102400 00054'024406 00055'066034 00056'061034 DSZ JMP SUB LDA DOB DOA COUNT DELAY 0,0 1,CHLMP 1,34 0,34

					C-28	3
00057 060177		INTEN				
00060.0060025		JSR	e.FRET			
00061 003777	MAX:	3777	JMAX VOL	TAGE IS	5 VOLTS	
00062.000005	CHLMP:	2	I LAMP C	HANNEL I	S #2	
00063 050000	DEL:	50000	JAPPROX.	0.15 SE	C DELAY (LAMP	ON)
00064 000000	COUNT:	Ø				
	HANG O	N UNTIL	BUTTON VO	LTAGE		
	JIS LES	S THAN 2	.5 VOLTS			
00065 020422	BACK:	LDA	Ø,CH3	INO LONG	ER CHANNEL Ø	
00066'061041		DOA	Ø, DVCE			
00067 063641		SKPDN	DVCE			
00070 000777		JMP	1			
00071 060441		DIA	Ø, DVCE			
00072'024412		LDA	1,01000			
00073'106512		SUBL#	Ø,1,SEC			
00074'000771		JMP	BACK			
00075'000712		JMP	LOOP			
00076'024412	HIT:	LDA	1,MASK			
00077 060510		DIAS	Ø,TTI			
00100'123400		AND	1,0 Ø, eN+2,3			
00101'043613 00102'060177		STA	ØJENT233			
00103 0060025		JSR	e.FRET			
00104'001000	C1000:	1000	eernet			
00105'000020	CH1:	20				
00106'000040	CH2:	40				
00107 000060	CH3:	60				
00110'000177	MASK:	177				
		.END				

```
001 C----- SECOND OVERLAY------
002 C--ROUTINE TO BUILD BLOCKS FROM LINES
          COMMON KEY(256), IBLOC(1536), IDUM(608), I1(768),
003
          I2(768), LIST(32), LISTC(128), IX(512), IY(512)
004
       *
          COMMON/HANDY/N,L,IACC
005
006 C
          N=NUMBER OF POINTS
007 C
          L=NUMBER OF LINES
008 C
.009 C
          N=LIST(1)
010
          L=LIST(2)
IACC=LIST(3)
Ø11
012
          IF(L.LE.2) GOTO 18
013
          PI=4.0*ATAN(1.0)
014
          PI2=2.0*PI
015
          PI05=0.5*PI
016
          PI180=PI/360.
017
          LBIT=100000K
018
          MASK=77777K
019
020 K=1
021 NBLOC=0
022 C--SET FLAGS ON ALL LINES--
          DO 1 LL=1,L
I1(LL)=I1(LL).OR.LBIT
023
024
          I2(LL)=I2(LL).OR.LBIT
025
      1
026 C--FIND IF ANY FLAGS STILL LEFT--
                      PRINTER LAKES () / TABLE
          DO 3 LL=1,L
027
      2
          IF(II(LL) . AND . LBIT) GOTO 4
628
          IF(12(LL).AND.LBIT) GOTO 5
029
          CONTINUE
030
      3
          IF(NBLOC.GT.Ø) GOTO 17
031
          CALL OVLAY(1,KEY)
      18
032
033
          PAUSE
          GOTO 18
034
          KEY(NBLOC+1)=K JALL FLAGS MUST BE DOWN .
      17
035
          CALL CHARO(135) ;FIND CENTROIDS ETC.
036
          CALL CENT(NBLOC)
I1(LL)=I1(LL).AND.MASK
037
038
       4
          IEND1=I1(LL)
039
          IEND2=I2(LL) . AND . MASK
040
          GO TO 6
041
          I2(LL)=I2(LL) .AND .MASK
042
       5
043
                       J (FLAG MUST ALREADY BE DOWN)
          IEND2=I1(LL)
044
       6
          ISTART=IENDI
045
          IPNT=1
046
          LISTC(1)=LL
047
          GAMSUM=0.0
048
```

 049
 IXD=IX(IEND2)-IX(IEND1)

 050
 IYD=IY(IEND2)-IY(IEND1)

 051
 IF(IXD.NE.0) GOTO 8

 052
 IF(IYD.LT.0) GOTO 7

 053
 ALFOLD=PI/2.0

 054
 GOTO 9

 055
 7

056		GOTO 9	
Ø57	8	ALFOLD=ATAN (ABS(FLOAT(IYD)/FLOAT(IXD)))	
058		1F(IXD+LT+0) GOTO 10	
059		IF(IYD.GT.0) GOTO 9	
060		ALFOLD=PI2-ALFOLD	
061		GOTO 9	
062	10	IF(IYD.GT.0) GOTO 11	
063	THE PARTY OF	ALFOLD=ALFOLD+PI	
064		GOTO 9	
065	11	ALFOLD=PI-ALFOLD	
866	CFIND	MOST CLOCKWISE LINE FROM LL	
067	9	LMAX=0	
068		GAMAX=PI	
069		DO 12 LIN=1.L	
070		IF(LIN.EQ.LL) GOTO 12	
Ø71		IF(II(LIN) . AND . LBIT) GOTO 13	
072	16	IF(I2(LIN).AND.LBIT) GOTO 14	
073		GOTO 12	
074	13	IF((II(LIN).AND.MASK).NE.IEND2) GOTO 16	
075		IE1=IEND2	
076		IE2=I2(LIN) . AND . MASK	
077		GOTO 15	
078	14	IF((I2(LIN) . AND . MASK) . NE . IEND2) GOTO 12	
879		IE1=IEND2	
080		IE2=II(LIN) . AND . MASK	
081	15	IXD=IX(IE2)-IX(IE1)	
082		IYD=IY(IE2)-IY(IE1)	
083		IF(IXD.NE.0) GOTO 20	
084		IF(IYD.LT.Ø) GOTO 19	
085		ALF=PI/2.0	
086	10.1001	GOTO 22	
087	19	ALF=1.5*PI	
088		GOTO 22	
089	20	ALF=ATAN(ABS(FLOAT(IYD)/FLOAT(IXD)))	
090		IF(IXD.LT.0) GOTO 21	
091		IF(IYD.GT.0) GOTO 22	
092		ALF=PI2-ALF	
093		GOTO 22	
094	21	IF(IYD.GT.0) GOTO 23	
095		ALF=ALF+PI	
096	00	GOTO 22 ALF=PI-ALF	
097 098	23		
099	66	GAM=ALF-ALFOLD IF(GAM.GE.PI) GAM=GAM-PI2	
100		IF (GAM.LTPI)GAM=GAM+PI2	
101		IF (GAM.GE.GAMAX) GOTO 12	
102		GAMAX=GAM JMOST CLOCKWISE ANGLE YET	
103		LMAX=LIN J WITH ITS CORRESPONDING LINE .	
104		ALFMAX=ALF	
104			

105 IED1=IE1 106 IED2=IE2 107 12 CONTINUE 108 IF(LMAX.EQ.0) GOTO 28 JDEAD END ! 109 C--KNOCK DOWN FLAG FOR THAT LINE--110 IF((I1(LMAX).AND.MASK).EQ.IED2) GOTO 24

111		I1(LMAX)=IED1	
112		GOTO 25	
113	24	I2(LMAX)=IED1	
114	25	GAMSUM=GAMSUM+GAMAX JSUM OF ALL BLOCK ANGLES	
115		IPNT=IPNT+1 JPOINTER TO TEMP. LIST OF LINES	
116		LISTC(IPNT)=LMAX	
117		IF(IED2.EQ.ISTART) GOTO 26	
118		IL MAY INCULINE DECOMES OLD LINE	
119			
120		IENDO-IEDO	
121		0 0 1 0 0	
122	26	TERCANCIN CT & ALCOTO O	
123		UDI CO UDI COLI	
124		KEY (NBLOC) =K	
1 1 1 1 1 1 1 1 1 1 1 1 1 1 1 1 1 1 1	CTHE	NEXT SECTION MERGES ADJACENT LINES IF	
126	Contraction of the second s		
127			
128			
129			
130		IP1=I1(LINE) . AND . MASK	
131			
132	31		
133	32	IX1=IX(IP1)	
134			
135		IX0=IX(ISTART)	
136		IYØ=IY(ISTART)	
137		IXD=IX1-IXØ	
138		IYD=IY1-IY0	
139		IF(IXD-E0.0) GOTO 43	
140		ALF1=ATAN2(FLOAT(IYD),FLOAT(IXD))	
141		GOTO 44	
142	43		
143	44	ALF1R=ALF1	
144		DO 50 IK=2, IPNT	
145		IF(IK.EQ.IPNT) GOTO 51	
146		LINE=LISTC(IK)	
147		IF(IP1.EQ.II(LINE)) GOTO 41	
148		IP2=I1(LINE) . AND . MASK	
149	- Contact	GOTO 42	
150	41	IP2=I2(LINE) . AND . MASK	
151	42	IX2=IX(IP2)	
152		IY2=IY(IP2)	
153	47	IXD=IX2-IX1	
154		IYD=IY2-IY1	
155		IF(IXD.E0.0) GOTO 45	
156		ALF2=ATAN2(FLOAT(IYD),FLOAT(IXD))	
157		GOTO 46	
158	45		
159	46	IF(ABS(ALF2-ALF1).LT.PI180) GOTO 53	
160		IBLOC(K)=IP1	
161		K=K+1	
162		IP1=IP2	
163		ALF1=ALF2	
164		IX1=IX2	
165		IY1=IY2	

166		GOTO 50	
167	51	IX2=IX(ISTART)	
168		IY2=IY(ISTART)	
169		GOTO 47	
178	53	IP1=IP2	
171	50	CONTINUE	
0.000		LINE TO DO NOW	
173		IF(ABS(ALFIR-ALF1).LT.PI180) GOTO 48	
174		IBLOC(K)=ISTART	
175		K=K+1	
176	48	IF(K-KEY(NBLOC).GT.2) GOTO 52	
10000	CWEED		
178		K=KEY(NBLOC)	
179		NBLOC=NBLOC-1	
180		GOTO 2	
181	52	KI=KEY(NBLOC)	
182		K2=K-1	
183		CALL PLOTS(0, IX(IBLOC(K2)), IY(IBLOC(K2)))	
184		DO 49 KB=K1,K2	
185	49	CALL PLOTS(1, IX(IBLOC(KB)), IY(IBLOC(KB)))	
186		GOTO 2	
187	CDEAL	WITH DEAD END	
188	28	II(LL)=II(LL) . AND . MASK	
189		12(LL)=12(LL) . AND . MASK	
190		IF(IPNT.LE.I) GOTO 2	
191		IPNM=IPNT-I	
192		ITO=ISTART	
	CRESTO	DRE FLAGS TO PRECEEDING LINES	
194		DO 30 IL=1, IPNM	
195		LINE=LISTC(IL)	
196		IF(ITO.EQ.II(LINE)) GOTO 33	
197		TIU-TITLINET AND MASK	
198 199		I2(LINE)=I2(LINE) + OR + LBIT	
200	33	GOTO 30 ITO=I2(LINE).AND.MASK	
201	35	II(LINE)=II(LINE) +OR+LBIT	
282	30	CONTINUE	
203	.50	6010 9	
204		END	
		END AND AND AND AND AND AND AND AND AND A	

001		SUBROUTINE CENT (NBLOC)		
002	CTO	FIND THE AREAS AND CENTROIDS OF ALL BLOCKS		
003	0 10	COMMON KEY(256), IBLOC(1536), LENG(1536), IAREA(256	52.	
		ICX(256), ICY(256), IX(512), IY(512)		
004		COMMON/HANDY/N,L,IACC		
005				
006		AMIN=IACC*IACC*5		
007		DO 1 N=1,NBLOC		
008		KI=KEY(N)		
009		K2=KEY(N+1)-1		
010	CFIN			
Ø11		IXM=1023		
012		IYM=780		
013		DO 3 K=K1,K2		
014		IP=IBLOC(K)		
015		IF(IX(IP).LT.IXM) IXM=IX(IP)		
016		IF(IY(IP).LT.IYM) IYM=IY(IP)		
017	3	CONTINUE		
018	CFIN	D BLOCK AREAS		
019		AREA1=0.0		
020		AREA2=0.0		
021		IP1=IBLOC(K2)		
022		DO 2 K=K1,K2		
023		IP2=IBLOC(K)		
024		IX1=IX(IP1)-IXM		
025		IX2=IX(IP2)-IXM		
026		IY2=IY(IP2)-IYM		
027		IY1=IY(IP1)-IYM		
Ø28		AREA1=AREA1+FLOAT(IX2-IX1)*FLOAT(IY1+IY2)/2.0		
029		AREA2=AREA2+FLOAT(IY2-IY1)*FLOAT(IX1+IX2)/2.0		
030	2	IP1=IP2		
031	-	AREA= (AREA1-AREA2)/2.0		
032		IF (AREA.LE.AMIN) GOTO 13		
033		IAREA(N)=AREA/AMIN		
034	CNOW	FIND MOMENTS OF AREAS ABOUT IXM. IYM		
and the second second	CIVOW			
035		XM=0.0		
036		YM=0.0		
037		IP1=IBLOC(K2)		
038		DO 12 K=K1,K2		
039		IP2=IBLOC(K)		
040		IX1=IX(IP1)-IXM		
041		IX2=IX(IP2)-IXM		
042		IY1=IY(IP1)-IYM		
043		IY2=IY(IP2)-IYM		
044		F1=FLOAT(IX2-IX1)/2.0		
045		F2=FLOAT(IX2+IX1)		
046		IF(IY2-IY1) 5,6,7		
040	6	XM=XM+F1*F2*FLOAT(IY1)		
	0	GOTO 8		
048	-	XM=XM+F1*(F2*FLOAT(IY2)+FLOAT(IY1-IY2)*FLOAT(2*I	X1+1X	2)/3.0)
049	5	AM-AMTER TECHNICITZITELUMICITI-TTZITELUMICZ+1		

	0010 0
7	XM=XM+F1*(F2*FLOAT(IY1)+FLOAT(IY2-IY1)*FLOAT(IX1+IX2*2)/3.0)
8	G1=FLOAT(IY2-IY1)/2.0
	G2=FLOAT(IY2+IY1)
	IF(IX2-IX1) 9,10,11
10	YM=YM-G1*G2*FLOAT(IX1)
	7 8 10

056			G0T0 12		
057		9	YM=YM-G1*(G2*FLOAT(IX2)+FLOAT(IX1-IX2)*FLOAT(IY	2+2*111)/3.0)
058			GOTO 12		
059		11	YM=YM-G1*(G2*FLOAT(IX1)+FLOAT(IX2-IX1)*FLOAT(IY	1+2*IY2)/3.0)
060		12	IP1=IP2		
Ø61			ICX(N)=IFIX(XM/AREA+0.5)+IXM		
062			ICY(N)=IFIX(YM/AREA+0.5)+IYM		
063			CALL CROSS(ICX(N), ICY(N))		
064			GOTO 1		
065		13	IAREA(N)=0.0		
066		1	CONTINUE		
067	C	·TO	COMPUTE THE LENGTHS OF EACH EDGE		
068			DO 80 N=1, NBLOC		
069			K1=KEY(N) K2=KEY(N+1)-1		
Ø70 Ø71			IPA=IBLOC(K2)		
072			KN=K2		
073			DO 81 K=K1+K2		
074			IPB=IBLOC(K)		
075			XDIF=IX(IPB)-IX(IPA)		
076			YDIF=IY(IPB)-IY(IPA)		
077			LENG(KN)=SQRT(XDIF*XDIF+YDIF*YDIF) + 0.5		
078			KN=K		
079		81	IPA=IPB		
080		80	CONTINUE		
Ø81	C				
085		25	CALL CURS(ID, IXX, IYY)		
083			CALL CHARO(159) IF(ID.E0.197) GOTO 20 J"E" FOR "ERASE"		
084			IF(ID.E0.197) GOTO 20 J"E" FOR "ERASE" IF(ID.E0.200) GOTO 30 J"H" FOR "HARD COPY"		
Ø85 Ø86			IF(ID.E0.208) GOTO 50 ;"P" FOR "PHASE"		
087			IF(ID.EQ.193) GOTO 22 ;"A" FOR "ALL"		
088			IF(ID.EQ.211) GOTO 60 J"S" FOR "SINGLE"		
089			IF(ID.EQ.210) GOTO 70 J"R" FOR "RESTORE"		
090			GOTO 25		
091		20	DO 24 NEL-NELOC		
092			IF(IABS(ICX(N)-IXX).GT.IACC) GOTO 24		
093			IF(IABS(ICY(N)-IYY).GT.IACC) GOTO 24		
094			IF(IAREA(N).LE.Ø) GOTO 24		
095			IAREA(N)=-IAREA(N)		
096			GOTO 22		
097		24	CONTINUE		
098			6010 25		
099		55	CALL CHARO(155)		
100			CALL CHAROLINE?		
101			DO LI M-ISHOLOO		
102			IF(IAREA(N).LE.0) GOTO 21 K1=KEY(N)		
104			K2=KEY(N+1)-1		
105			CALL PLOTS(0, IX(IBLOC(K2)), IY(IBLOC(K2)))		
106			DO 23 K=K1,K2		
107		23	CALL PLOTS(1, IX(IBLOC(K)), IY(IBLOC(K)))		
108			CALL CROSS(ICX(N),ICY(N))		
109		21	CONTINUE		
110			GOTO 25		

111	30	CALL COPY (ISWIT) JCHECK FOR SWITCH
112		IF(ISWIT .EQ. Ø) GO TO 25
C N L CONTRACT		
113		DO 31 N=1.NBLOC
114		IF(IAREA(N).LE.0) GOTO 31
115		K1=KEY(N)
116		K2=KEY(N+1)-1
117		11=1X(IBLOC(K2))*4-2047
118		I2=IY(IBLOC(K2))*4-2047
119		CALL PLOT(II, 12, 3)
120		DO 32 K=K1,K2
121		I1=IX(IBLOC(K))*4-2047
122		12=1Y(IBLOC(K))*4-2047
123	32	CALL PLOT(11,12,2)
	56	IC1 = ICX(N) * 4
124		
125		IC2=ICY(N)*4
126		CALL PLOT(IC1-2087, IC2-2047, 3)
127		CALL PLOT(IC1-2007, IC2-2047, 2)
128		CALL PLOT(IC1-2047, IC2-2087, 3)
129		CALL PLOT(IC1-2047, IC2-2007,2)
130	31	CONTINUE
	51	CALL PLOT (-2047,-2047,3)
131		
132	1.1.040	GOTO 25
133	40	CALL CHARO(155)
134		CALL CHARO(140)
135		CALL OVLAY(1,KEY)
136		GOTO 25
137	50	CALL CHARI(IN)
and the second second	50	IF(IN.EQ.177) GOTO 40 J"1" FOR "PHASE 1"
138		
139		IF(IN.NE.179) GOTO 25 J"3" FOR "PHASE 3"
140		CALL CHARO(155)
141		CALL CHARO(140)
142		IBLOC(1536)=NBLOC
143		CALL OVLAY (3, KEY)
144		GOTO 25
and the second second	10	
145	60	DO 61 N=1,NBLOC
146		IF(IABS(ICX(N)-IXX).GT.IACC) GOTO 61
147		IF(IABS(ICY(N)-IYY).GT.IACC) GOTO 61
148		GOTO 62
149	61	CONTINUE
150		GOTO 25
151	62	NN=N
152		IF(IAREA(NN).LE.0) GOTO 25
153		CALL CHARO(155)
154		CALL CHARO(140)
155		K1=KEY(NN)
156		K2=KEY(NN+1)-1
157		CALL PLOTS(0, IX(IBLOC(K2)), IY(IBLOC(K2)))
158		D0 63 K=K1,K2
159	63	CALL PLOTS(1, IX(IBLOC(K)), IY(IBLOC(K)))
	6.6	CALL CROSS(ICX(NN), ICY(NN))
160		
161		CALL CHARI(IN)
162		IF(IN.NE.197) GOTO 22
163		IAREA(NN)=-IABS(IAREA(NN))
164		GOTO 22
165	70	DO 71 N=1.NBLOC
105		

166 167 168 169 170	IF(IAREA(N).GE.Ø) GOTO 71 IAREA(N)=IABS(IAREA(N)) 71 CONTINUE GOTO 22 END	

List of Phase 3 Global Symbols

Symbol Name	Originating Routine	Purpose of Symbol
CONTR	CONTR	Iteration and Control routine entry
FEET	INPUT	ASCII Length Descriptor
MOVFL	INPUT	Memory overflow message
MU	FORD	Default value of friction coefficient
OPTIN	CYCLE	Pointer to option input routine
POUND	INPUT	ASCII force descriptor
PUP	REBOX	Pressure segment test entry
TRANS	TRANS	Initial translation routine entry
.ALLB	UPDAT	Pointer to routine to update all blocks
.ALPH	UTIL	Pointer to routine to set Tektronix in alpha node
.AXIS	UTIL	Pointer to routine to draw axes on screen
.BSIZ	TRANS	Number of words in block data arrays, excluding corners
.0100	CONTR	A constant (=100 octal)
.CHEK	UTIL	Pointer to routine check if character is a digit
.CLNC	TAPE	Pointer to tape checking routine
.CPNT	UPDAT	Pointer to word that can be changed
.CURS	TEK	Pointer to routine that enables cursor
.DBØ	UTIL	Pointer to Decimal to Binary conversion routine
.DBIN	UTIL	Pointer to Decimal to Binary conversion routine
.DCM	MOUIT	Pointer to routine to move a fixed block
.DISB	DISPL	Pointer to routine that plots a single block
.DISP	DISPL	Pointer to routine that plots all blocks on paper
.DISS	DISPL	Pointer to routine that plots all blocks on screen
. DMBN	INPUT	Block number of fixed block to be moved
.DMBP	INPUT	Block data pointer of fixed block to be moved
.EMPT	TRANS	Head of empty list
.FORD	FORD	Pointer to force/displacement routine
.GETT	UTIL	Pointer to routine to accept keyboard character
.HEAV	LOADS	Pointer to routine to modify block weights

.HITCUTILPointer to routine to detect cursor hit on block.HITSHITSPointer to routine to detect cursor hit on edge.IACCUTILAccuracy limit for hits on centroids

rates an entropy of an entropy of the second

. INP	INPUT	Pointer to friction input routine
. IPRN	UTIL	Pointer to binary to decimal conversion routine
.KET	CYCLE	Pointer to routine to calculate kinetic energy
.LENG	UTIL	Pointer to routine to return length of an edge
.LODE	INPUT	Pointer to routine for numerical applied load input
.LPAP	CONTR	Flag for hard copy load plot option
.LPLS	DISPL	Pointer to routine for plotting loads on screen
.M1	TRANS	Pointer to start of block data pointers
.M2	TRANS	Pointer to start of block data arrays
.M3	TRANS	Pointer to start of boxes
.M4	TRANS	Pointer to start of linked lists of block corners
.M5	TRANS	Pointer to start of block pointers to contact lists
.M6	TRANS	Pointer to start of linked list area
.M7	TRANS	Pointer to start of free memory
.MEM	TRANS	Highest memory location
.MESS	UTIL	Pointer to routine that prints messages on screen
.MFLG	INPUT	Flag for displacement control option
.MOT	MOTIO	Pointer to law of motion routine
.MOVE	INPUT	Pointer to input routine for moving fixed block
.MSKR	REBOX	A constant (377 octal)
.NUM	TRANS	Total number of blocks
.NVEC	DISPL	Flag for printing vector magnitudes
.OVL	TAPE	Pointer to routine to read first overlay
.PAGE	UTIL	Pointer to routine that clears the screen
.PEMT	INPUT	Head of pressure segment empty list
.PFLG	CONTR	Flag to control plotting when running
.PLTS	TEK	Pointer to line drawing routine entry
.PON1	PONT	Pointer to routine that returns global coordinates
.PON2	PONT	Pointer to quick entry to above routine
.PRES	INPUT	Head of pressure segment list
.PRN1	UTIL	Pointer to routine that prints a single character

.PRN2	UTIL	Pointer to routine that prints character in ACD
.PSEG	INPUT	Pointer to pressure segment input routine
.PSIZ	TRANS	Number of words in each contact entry
.READ	TAPE	Pointer to routine to read a stored data set
.REBX	REBOX	Pointer to re-boxing routine entry
.REBZ	REBOX	Pointer to re-boxing routine, alternate entry

		C-39
.RLNC	TAPE	Pointer to tape reading routine
.ROT	MOTIO	Constant of integration for angular velocity
.RSET	CYCLE	Pointer to routine that resets cycle counter
.SCAL	UTIL.	Pointer to vector scaling routine
.SING	UPDAT	Pointer to single block updating routine
.SPRP	INPUT	Pointer to beginning of friction table
.STEP	CYCLE	Pointer to routine to increment cycle counter
.SYCL	INPUT	Frequency of movement of fixed block
.TIME	FORD	Pointer to routine to change time step
.TPRN	CYCLE	Pointer to routine that displays cycles
.TREC	MOTIO	Inverse time step
.TYP	UTIL	Pointer to return surface type number for edge
.UD	INPUT	Unit of displacement
.UINP	INPUT	Pointer to units input routine
.UREP	CONTR	Update frequency
.UW	INPUT	Unit weight
.VEC	CONTR	Vector plotting flag
.VFAC	UTIL	Vector scaling factor
.WLNC	TAPE	Pointer to tape writing routine
.WORD	UTIL	Pointer to routine to get alphanumeric string
.WRIT	TAPE	Pointer to routine to store a data set
.XCGD	INPUT	X - component of fixed block displacement
.YCGD	INPUT	Y component of fixed block displacement

 		1 mar 1 mar 1 1				-
		•TITL				C-40
		ATE NEW	DATA STI		FROM	
	THE OR		ORTRAN A			
		and the second sec			•M3. •NUM.	
		.ENT		5. • M6. • M	7. • EMPT. •	PSIZ
		• ENT	-MEM			
		.EXTN	CONTR			
		•EXTD	Property of the local data and t		LLB, DISS	MSKR
		.EXTD	.OVL N	1ESS. TP	RN	
	North States	. ZREL				
00000-000000	.MEM:	Ø	JHIGHES	ST MEMOR	Y LCTN	
00001-000000	• M 1 =	Ø				
00002-000000	•M2:	Ø				
00003-000000	•M3:	Ø				
00004-000000	•M4:	Ø		ARRAY ST.		
00005-000000	•M5:	Ø	JLINK A	ARRAY EN	D+1	
00006-000000	•M6:	Ø				
00007-000000	•M7:	Ø			E LOCATIO	N
00010-000000	• EMPT:	0		STUDIES CO.	ST START	
00011-000014	•PSIZ:	14		INTRY SI	ACC 2014	
00012-000000	-NUM:	Ø		OF BLO	and the second se	
00013-000025	.BSIZ:	25	ISTARI	OF POIN	DATA	
000001000000		.NREL	FORTDA		LOCATIO	NC
00000.000000	AREA:	0	JFURIRA	an commu	N LOCATIO	ND
00001'000000	ICX:	Ø				
00002'000000	ICY:	0				
00003'000000	KEY:	0				
00004'000000	LENG:	Ø		PROCRA		
00005'000404	NMAX:	404	JIOP OF	PROGRAM	AREA	
00006.000400	F400:	400				
00007'000417'	NEXTR:	NEXT	10			
000012		.RDX	10 S MUST E			
			S CHANGE	State and state and state	and the second s	
			MS, PHAS			
00010.000011.		+1	MOJ FRAS	23100	- 100	
00011 001001	IDL.	513	JIY	1		
00012'001000		512	IX	;		
00013'000400		256	FICY	1		
00014'000400		256	JICX			
00015'000400		256	JIAREA) FOR	. ARRAY	NAMES
00016'003000		1536	JLENG)	· Annai	MANED
00017 003000		1536	JIBLOC	5		
00020 000400		256	JKEY	;		
00020 000400	3	250	JUE I			
00021 177770	COUNT:	-8	IMINUS	NO. OF	RRAYS	
000010	0000000	. RDX	8			
00022'001000	STEP:	1000	Contraction of the			
00023'100600	HIGH:	77600+1	000	JALLOWS	5 200 WDS	FOR LDR
00024'000303'	IPXR:	IPX				and server
00025'000304'	IPYR:	IPY				
00026.000000	IBLOC:	ø				
	Contraction of the local distance	80.0				

TRANS: 3.TBL 00027 034761 LDA 2,COUNT 00030'030771 LDA 00031'126400 SUB 1.1 JTO FIND TOTAL COMMON BLOCK SIZE 0.0.3 SUM: 00032'021400 LDA 00033'107000 0.1 ADD 2.2 00034'175400 INC

3

00035'151404	INC	2,2,52R	C-41
00036'000774	JMP	SUM	the second se
	JCOMMON SIZE IN	AC1	
	INOW SIZE CORE		
00037 020763	LDA	Ø,STEP	
00040'034763	LDA	3,HIGH	
00041 116400	SUB	0.3	
00042'055777	STA	3,-1,3	
00043'031777	LDA	2,-1,3	
00044'156414	SUB#	2,3,52R	
00045'000774	JMP	4	
00046'050000-	STA	2MEM	
	HIGHEST USEABL	E MEMORY	IS IN AC2
00047'132400	SUB	1,2	ILOWEST LOC. OF COMMON
00050 050733	STA	2.KEY	
	JCOMPUTE LOCATI	ONS OF I	NDIVIDUAL ARRAYS
00051 '024747	LDA	1, TBL+1	0
00052'133000	ADD	1,2	
00053'050753	STA	2, IBLOC	
00054'024743		1, TBL+7	
00055'133000	ADD	1,2	
00056'050726	STA	2.LENG	
00057 024737	LDA	1, TBL+6	
00060 133000	ADD	1,2	
00061 050717	STA	2. AREA	
00062'024733	LDA	1.TBL+5	
00063'133000	ADD	1,2	
00064'050715	STA	2,ICX	
00065'024727	LDA	1, TBL+4	
00066'133000	ADD	1,2	
00067 '050713	STA	2, ICY	
00070 024723	LDA	1. TBL+3	
00071 133000	ADD	1,2	
00072 052732	STA	2, BIPXR	
00073 024717	LDA	1, TBL+2	
00074'133000	ADD	1,2	
00075 052730	STA	2, eIPYR	
00076'030706	LDA	2.LENG	
00077'021377	LDA	0,-1,2	
00100'040012-	STA	Ø NUM	INUMBER OF BLOCKS
00101'101005	MOV	0,0, SNR	
00102 0060065	JSR	e.OVL	JEXIT NO BLOCKS
00103'022702	LDA	0. PNMAX	ISET UP START OF DATA AREA
00104'040001-	STA	Ø M1	
00105'024701	LDA	1.F400	
00106'123000	ADD	1,0	
00107 '040002-	STA	Ø M2	
00110'102400	SUB	0.0	INITIALIZE COUNTERS
00111'040566	STA	Ø.NB	
00112'040566	STA	0.NP	
00113'034001-	LDA	3++M1	JINITIALIZE POINTERS
00114'054566	STA	3, PPNT	
00115'030002-	LDA	2M2	

00116'050563 STA 2,8PNT 00117'051400 STA 2,0,3 FIRST BLOCK POINTER INSTALLED

00120'034660 BACK: LDA 3,AREA 00121'024556 LDA 1,NB 00122'137000 ADD 1,3 00123'021400 LDA 0,0,3

JGET AREA, BLOCK NB

				C 40
00124'101004		MOV	0,0,52R	C-42
00125'101112		MOVL#	0,0,SEC	
00126'002661		JMP	ENEXTR	INEGATIVE, OR ZERO, AREA
	3	CTA	0.14.0	STORE AREA
00127'041014		STA	0,14,2	INITIALIZE THE FOLLOWING:
00130'102400		SUB	0,0	FINITIALIEE THE FOLLOWING.
00131'049562		STA	Ø,MAX	JLOW X
00132'041002		STA	0,2,2	JLOW Y
00133'041004		STA	8,4,2	
00134'041011		STA	0,11,2	(SIN)
00135'041005		STA	0,5,2	X-VEL
00136'041006		STA	0,6,2	JALPHA-DOT
00137 041012		STA	0,12,2	JLOW ALPHA
00140'041007		STA	0,7,2	XFSUM
00141 041015			0,15,2	JY-VEL
00142'041016		STA	0,16,2	JYFSUM
00143'041017		STA	0,17,2	INSUM
00144'041020		STA		JDELTA-X
00145'041021		STA	0,21,2	JDELTA-Y
00146'041022		STA	0,22,2	JDELTA-ALPHA
00147 041023		STA	0,23,2	JX LOAD
00150'041024		STA	0,24,2	JY LOED
00151'100000		COM	0,0	MOOCH - NEADECT TUINC TO I
00152'041010		STA	0,10,2	; (COS) = NEAREST THING TO 1
	;			
00153'034626		LDA	3.ICX	
00154'137000		ADD	1,3	LOT TOWNER
00155'021400		LDA	0,0,3	JGET ICX(NB)
00156'041001		STA	0,1,2	PUT IN NEW BLOCK LIST
00157'040537		STA	Ø,IX	JTEMP STORE FOR LATER USE
00160'034622		LDA	3, ICY	
00161'137000		ADD	1,3	ART LOUGHT
00162'021400		LDA	0,0,3	JGET ICY(NB)
00163'041003		STA	0,3,2	JPUT IT AWAY
00164'040531		STA	ØJIY	JAS WITH IX
00165'034616		LDA	3.KEY	
00166'137000		ADD	1,3	NURVENIDA
00167 021400		LDA	0,0,3	;KEY(NB)
00170'025401		LDA	1,1,3	JKEY(NB+1)
00171'106400		SUB	0,1	NUMBER OF POINTS THIS PLOCK
00172'045000		STA	1,0,2	INUMBER OF POINTS THIS BLOCK
00173'024013-		LDA	1,.BSIZ	
00174'133000		ADD	1,2	
00175'126520		SUBEL	1,1	THEY (NON-1
00176'122400		SUB	and the second second second	IKEY(NB)-1
00177'034605		LDA	3.LENG	POINTER TO LENGTH ARRAY
00200'117000		ADD	0,3	
00201 054506		STA	3.FANG	
00202 054506		STA	3. FENG	
00203'034623		LDA	3,IBLOC	
00204'117000		ADD	0,3 3.5ING	
00205'054504		STA	3,FING	FOR LONG BLOCK
00206'054504		STA	3.FONG	PENDA COPT FOR LONG BLOCK

00207 021400 00210'122400 00211'034472 00212'117000 00213'025400 00214'034470

LOOP: LDA SUB LDA ADD LDA LDA

3

0,0,3 1,0 3, IPX 0.3 1,0,3 3, IPY SPOINT NUMBER 3P. NUM -1 **#POINTER TO X CO-ORD IN IPX** ■X CO-ORD IN AC1

-				
	00215'117000	ADD	0,3	POINTER TO Y CO-ORD IN AC3
	00216'020500	LDA	Ø,IX	JGET XC BACK
	00217'122400	SUB	1,0	JXC-XP (RELATIVE X, XR)
	00220'100400	NEG	0.0	
	00221 040465	STA	0.TEMP	
	00222'024463	LDA	1,0NE27	1127
	00223'101112	MOVL#	0,0,SEC	
	00224'100400	NEG	0.0	JABS(XR)
	00225'106512	SUBL#	0,1,SEC	; IS ABS(XR)>127 ?
	00226'000472	JMP	FWORD	JYES, TREAT AS LONG BLOCK
	00227 024464	LDA	1.MAX	
	00230 106512	SUBL#	0,1,SEC	
	00231 '040462	STA	Ø.MAX	
	00232 020454	LDA		JGET ACØ WITH CORRECT SIGN
	00233'0240055	LDA	1. MSKR	
	00234'123700	ANDS	1,0	MASK OFF LEFT BYTE, AND SWAP
	00235'025400	LDA	1,0,3	JY CO-ORD IN AC1
	00236'115000	MOV	0,3	FRETAIN XR IN LEFT BYTE OF AC3
	00237 020456	LDA	ØPIY	JGET YC BACK
	00240'122400	SUB	1,0	JYC-YP (RELATIVE Y, YR)
	00241 100400	NEG	0,0	TO CORRECT A BLUNDER !
	00242 040444	STA	ØTEMP	TO COMPLET A DECIDENT
	00243 024442	LDA		JDO AS WITH X
	00244'101112	MOVL#	0,0,SZC	JUG AS WITH ATT
	00245'100400	NEG	0,0	
	00246'106512	SUBL#	ØJ1JSZC	
	00247 000451	JMP		MUST BE LONG BLOCK
	00250 024443	LDA	1.MAX	MUST BE LONG BLOCK
			Ø,1,SEC	
	00251'106512	SUBL#		
	00252'040441	STA	Ø,MAX	
	00253 020433	LDA	Ø, TEMP	
	00254'0240055	LDA	1. MSKR	MARY OFF LEFT DATE
	00255'123400	AND	1,0	MASK OFF LEFT BYTE
	00256'163000	ADD	3,0	STORE FULL WORD IN LIST
	00257 041000	STA	0,0,2	STORE FULL WORD IN LIST
	00260'034427	LDA	3.FANG	ACET LENCTU OF STOP NO
	00261'021400	LDA		GET LENGTH OF SIDE NP
	00262'041001	STA	0,1,2	STORE LENGTH IN 2ND WORD
	00263'010415	ISZ	NP	
	00264 020414	LDA	Ø,NP	LOST HAY DOINTS
	00265 026414			JGET MAX POINTS
	00266'151400	INC	2,2	BUMP POINT POINTER
	00267 151400	INC	2,2	IC ND & MAYD 2
	00270'122513		14. C	SIS NP > MAXP ?
	00271 000507	JMP	OUT	TYES, END OF POINT LOOP
	00272'010417	ISZ	FING	INO, CARRY ON
	00273'010414	ISE	FANG	POINTER TO IDLOG APPAY
	00274'034415	LDA	3.FING	SPOINTER TO IBLOC ARRAY
	00275'126520	SUBEL	121	ADDUND ACAIN HE CO
	00276'000711	JMP	LOOP	ROUND AGAIN WE GO

00277'0000000 N 00300'0000000 N 00301'0000000 E 00302'0000000 F 00303'035600 1 00304'035600 1 00305'000177 0 00306'000000 T 00307'0000000 F

 NB:
 Ø

 NP:
 Ø

 BPNT:
 Ø

 PPNT:
 Ø

 IPX:
 35

 IPY:
 36

 ONE27:
 17

 TEMP:
 Ø

 FANG:
 Ø

3

Ø 3560Ø 36600 177

FORTRAN POINT ARRAYS

				C 44
00310.000000	FENG:	Ø		C-44
00311 000000	FING:	Ø		
00312.000000	FONG:	Ø		
00313 000000	MAX:	0		
00314'000000	SAVE:	Ø		
00315.000000	IY:	Ø		
00316'000000	IX:	0		
00317 020000	LBIT:	020000	ILONG BI	LOCK FLAG
	3			
	ITHIS S	ECTION U	SED WHEN	LONG BLOCKS ARE FOUND
00320'102400	FWORD:	SUB	0.0	
00321 040757		STA	Ø,NP	IRESTORE POINT COUNTER
00322'024757		LDA	1.BPNT	
00323 030013-		LDA	2, .BSIZ	START OF POINT DATA
00324'133000		ADD	1,2	JRESTORE POINT POINTER
00325'034765	LOOPL:	LDA		POINTER TO IBLOC ARRAY START
00326'126520		SUBEL	1.1	
00327'021400		LDA	0,0,3	POINT NUMBER
00330'122400		SUB	1,0	PNUM-1
00331 '034752		LDA	3, IPX	and the second of the second se
00332'117000		ADD	0.3	JPOINTER TO X CO-ORD IN AC3
00333'025400		LDA	1,0,3	JX CO-ORD IN AC1
00334'034750		LDA	3, IPY	In or one in nor
00335'117000		ADD	0.3	POINTER TO Y CO-ORD IN AC3
00336'020760		LDA	Ø,IX	JGET XC BACK
00337 106400		SUB	0,1	JXP-XC (RELATIVE X, XR)
00340'045000		STA	1,0,2	STORE XR IN LIST
00341 125112		MOVL#	and the second se	JTO RECORD MAX DIMENSION
00342'124400		NEG	1,1	TO RECORD THAT DIRECTOR
00343'020750		LDA	Ø.MAX	
00344'122512		SUBL#	1,0,SEC	
00345 044746		STA STA	1.MAX	
00346'151400		INC	2,2	BUMP POINT POINTER
00347 025400		LDA		JY CO-ORD
00350.020745		LDA	Ø.IY	JYC BACK
00351 106400		SUB	0,1	SYP-YC (RELATIVE Y, YR)
00352 045000		STA	1,0,2	PUT IT AWAY
		MOVL#	1,1,520	
00353'125112 00354'124400		NEG	1,1	
00355'020736		LDA	Ø.MAX	
		SUBL#	1,0,520	
00356'122512		STA STA	1,MAX	
00357 044734		INC	2,2	BUMP POINT POINTER
00360'151400		LDA	3.FENG	BOMP FOINT FOINTER
00361'034727				LENGTH SIDE NP
00362'021400		LDA STA	0,0,3	FLENGIN SIDE NF
00363'041000			2,2	
00364'151400		INC ISE	NP	
00365 010713				
00366'020712		LDA	0.NP	
00367'026712		LDA	1.9BPNT	
00370'122513		SUBL#	1,0, SNC	POTHT LIST DONE
00371'000404		JMP	OUTR	POINT LIST DONE
00372.010720		ISZ	FONG	
00373'010715		ISE	FENG	

00373'010715 00374'000731 00375'020722 OUTR: LDA 00376'107000 00377'046702 J

00400'102400 OUT:

SUB

0.0

LOOPL Ø,LBIT Ø,1 1,0BPNT JADD IN LONG BLOCK FLAG

S INCOME DATA

00401 040677		STA	Ø,NP	JRESET POINT COUNTER
00402'034677		LDA	3, BPNT	
00403 050676		STA	2, BPNT	
00404 010676		ISE	PPNT	
00405 052675		STA	2, PPPNT	
00406'102400		SUB	0.0	
00407 024704		LDÁ	1.MAX	
00410.0300055		LDA	2. MSKR	1>256 NOT ALLOWED
00411'132512		SUBL#	1,2,SEC	
00412'145000		MOV	2,1	
00413'131000		MOV	1,2	
00414'073301		MUL		
00415'045413		STA	1,13,3	JD*D (MAX) FOR M. OF I.
00416'030663		LDA	2, BPNT	
00417 010660	NEXT:	ISZ	NB	
00420'024012-		LDA	1 NUM	
00421 '020656		LDA	ØNB	
00422'122512		SUBL#		JIS NB>=NBLOC ?
00423'002435		JMP		INO, KEEP GOING
00424'102400		SUB	0.0	
00425'042655		STA		PUT ZERO ADDRESS IN LOCATOR LIS
00426 050003-		STA	2M3	INEXT FREE MEMORY
00420 030000	THE NE		Later of the second	ES ALL POINTS
	and the second se	RSE BOXE		
00427 024432		LDA	1,BOXSE	
00430'134400		NEG	1,3	
00431 147000		ADD	2,1	JLINK ARRAY START
00432 044004-		STA	1 M4	
00433'044432		STA	1, FREE	
00434'102000		ADC	0.0	
00404 100000	;NOTE:			NS END OF LIST.
00435'041000	PIG:	STA	2,0,0	SET ALL LINKS TO 17777
00436'151400		INC	2,2	3 INITIALLY
00437'175404		INC	3,3,SER	
00440'000775		JMP	PIG	
00441 102400		SUB	0.0	
00442'040420		STA		JBLOCK NUMBER
00443'034001-		LDA	3 M1	
00444'054422		STA	3.PPNTA	
00445'032421	AROUN:	LDA	2, PPNTA	
00446'151005	Altooni	MOV		SEND OF LIST?
00447'000465		JMP	DONE	;YES
00450 021000		LDA	0.0.2	FIRST BLOCK WORD
00451 024420		LDA	1.MSKR	WINDI DECOM MOND
00451 024420		AND	1,0	JGET POINT COUNT ONLY
		STA	Ø,PCNT	; POINT COUNT
00453'040414		SUB	1,1	, tothi ocont
00454'126400 00455'044406		STA	1.NPA	RESET POINT COUNTER
		JSR	0.PON1	GET CO-ORDS OF FIRST POINT
00456'0060015			PLACE	JULI UU UNDU UN TANET TEAT
00457 000416	DAGUDA	JMP	FLACE	

00460'000120' 00461'000320 00462'000000 00463'000000 00464'000400 00465'000000 00466'000000 00467'000000 00470'000100 00471'000377

BACKR: BACK BOXSZ: 320 NBA: Ø NPA: Ø 400 PRODE: FREE: Ø PPNTA: Ø 0 PCNT: C100: 100 MSKR: 000377

BOX ARRAY SIZE (20*15 OCTAL)

PROD LOCATOR SIZE

6	and the first of the second se				C-46
	00472'000000	NY:	Ø		
	00473'024770	COW:	LDA	1,NPA	
	00474 9060025		JSR	e.PON2	JOUICK ENTRY
	00475'044775	PLACE:	STA	1.NY	JNOW PUT NX IN AC1
	00476'105000		MOV	0+1	INOW COMPUTE WHICH BOX
	00477 034003-		LDA	3113	JTHE POINT NX, NY SHOULD BE
	00500 030770		LDA	2,0100	JASSOCIATED WITH, AND PLANT A
	00501'102400		SUB	0.0	LINK TO IT IN THE BOX ARRAY.
	00502'073101		DIV		; INPUT: NX IN AC1
	00503'137000		ADD	1,3	JAC3=AC3+NX/100
	00504'102400		SUB	0.0	
	00505 024765		LDA	1.NY	
	00506'073101		DIV		
	00507'127120		ADDZL	1.1	
	00510'127120		ADDZL	1.1	
	00511'137000		ADD	1,3	JAC3=AC3+(NY/100)*20
	00512'021400		LDA	0,0,3	FIRST LINK (MAY BE 0)
	00513'030752		LDA	2.FREE	FREE SPACE POINTER
	00514'041001		STA	0,1,2	FUT OLD LINK IN 2ND WORD
	00515'051400		STA	2,0,3	PUT NEW LINK IN BOX ARRAY
	00516'024744		LDA	1,NBA	
	00517'020744		LDA	Ø,NPA	
	00520'101300		MOVS	0,0	
	00521'123000		ADD	1,0	COMPOSITE (NPA:NBA)
	00522'041000		STA	0,0,2	PUT IN 1ST WORD
	00523'151400		INC	2,2	
	00524'151400		INC	2,2	
	00525'050740		STA	2,FREE	JUPDATE FREE POINTER
	00526'010735		ISE	NPA	
	00527 '014740		DSZ	PCNT	JDONE IF PCNT=0
	00530'000743		JMP	COW	
	00531 010735		ISE	PPNTA	
	00532'010730		ISZ	NBA	
	00533'000712		JMP	AROUN	
	00534'030731	DONE:	LDA	2,FREE	
	00535'050005-		STA	2, .M5	INEXT FREE LOCATION
		INOW PRI	EPARE FO	R PROD L	IST
	00536'024726		LDA	1, PRODE	
	00537'134400		NEG	1,3	
	00540'147000		ADD	2,1	FPROD LIST START
	00541 '044006-		STA	1. M6	FIXED POINTER
	00542'044007-		STA	1 M7	IMOVING POINTER
	00543'102000		ADC	0.0	
	00544'040010-		STA	Ø EMPT	INOTHING IN EMPTY LIST
	00545'041000	ITR:	STA	0,0,2	SET ALL LINKS TO -1
	00546'151400		INC	2,2	
	00547'175404		INC	3,3,SZR	
	00550'000775		JMP	ITR	
	00551'0060105		JSR	e.TPRN	
	00552'0060045		JSR	e.DISS	DISPLAY ALL BLOCKS
	00553'0060075		JSR	e.MESS	
	00554'000561'		TEXT		
	000010		DOX	10	

000012 00555'177076 00556'000017 000010 00557'002401 00560'177777 00561 050040 00562'040510

. RDX -450 15 .RDX JMP CNTRL: CONTR TEXT:

HA

8 CONTRL .TXT * P

10

C-46

00563'042523 SE 00564'052040 T 00565'051110 HR 00566'042505 EE 00567'000000 * 000027'

.END TRANS

				C-48	
		.TITL	TEK		
	ITO PLO			TEKTRONIX SCREEN:	
	3				
	3	JSR E.P		AND A DESCRIPTION OF THE DESCRIPTION	
	; (PUT		FOR BEAM	OFF,	
	1		EAM ON.	AND COLORED AND THE AND PLAN	
	3		POINT PLO	and the second	
	INPUT	COLUMN STATES	X CO-ORD		
	1	AGI -	r co-orb	INAIC	
	TO GET	CURSOR	CO-ORDINA	ATES AND CHARACTER:	
	3				
	3	JSR e.C	URS		
	,	CHAR			
		X			
	JWHERE:				
	I	CHAR=AD	DRESS OF	WORD CONTAINING	
	1	onnan no	KEY CHAI		
	3	X =AD	DRESS OF	NORD WITH X CO-ORD,	
	3	Y =		•••• ¥ ••	
	3				
		•ENT •ZREL	• PLTS • • (CURS	
00000-000017.	PITS:	TPLOT			
00001-000150'					
00001 000100		.NREL			
00000'040416	CHIN:		Ø,CCACØ	SAVE ACO	
00001 '063610		SKPDN	TTI	ISKP IF CHAR READY	
00002'000777		JMP	• - 1		
00003'060510		DIAS		READ CHAR	
00004'043400		STA		SIDRE CHAR	
00005 020411		LDA		RESTORE ACO	
00006'001401		JMP	1,3		
00007 040407	CHOUI:	STA		SAVE ACO	
00010'063511		SKPBZ	-1	SKIP IF NOT BUSY	
00011'000777 00012'023400		LDA		JGET CHARACTER	
00013'061111		DOAS		SHEP CHARACTER	
00014'020402		LDA		RESTORE ACO	
00015'001401		JMP	1,3	CONTRACTOR OF CONTRACTOR OF	
00016'000000	CCACØ:	Ø		ITEMP FOR ACO	
00017 040525	TPLOT:	STA	Ø,TPTX	JX CO-ORD	
00020'044525		STA	1, TPTY	IY CO-ORD	
00021 021400		LDA		IMORE FROM CALL+1	
00022.040224		STA	Ø, TPMOD		
00023'054520		STA		DISAUE CALL ADDRESS	
00024'101015		MOV#		SKA IF NEO Ø	UCOTOD
00025'000405		JMP		;= 3 INITIALIZE AND DARK	VECTOR
00026'101113		MOVL#		ISKIP IF < Ø INOMAL BRIGHT VECTOR	
UUUCI UUUHUJ		Ulti	A A A A A A A A A A A A A A A A A A A		

00027 000405 00030'006511 00031'000130' 00032'006507 TPTDV: 00033'000127' 00034'020511 TPTNRM: LDA 00035'101112 00036'102400 00037 034477 00040'162513

JMP JSR US JSR GS MOVL# SUB LDA SUBL#

IPINKM INOMAL BRIGHT VECTOR ISET TO ALPHA **e**CHOUE

CHOUZ JDATK VECTOR

Ø, TPTY ;GED Y 0,0,SEC ; SKF IF + 0.0 IMARE Ø 3, D780 ; UPTER Y BOUND 3,0, SNC ISKT IF ON SCREEN

00041 161000		MOV	3,0	JSET TO EDGE	
00042'040503		STA	Ø, TPTY	JSAVE GOOD Y	
00043'101120		MOVEL	0.0	JUSE UPPER 5 BIT	S
00044'101120		MOVEL	0.0		
00045'101120		MOVEL	0.0		
00046'101300		MOVS	0.0	JAND SWAP HALVES	
00047 034463		LDA	3,8040	JHI Y TAG	
00050'163000		ADD	3,0	JPUT IN CHAR	
00051 040476		STA		PIUSE A TEMP	
00052'006467		JSR	eCHOUZ	SHIP HI Y 5	
00053'000147'		TPTTMP			
00054'020471		LDA	STATISTICS STATISTICS		
00055'034453		LDA	3,8037	JMASK	
00056'163400		AND	3,0	ILEAVE LOW Y 5	
00057 034455		LDA	3,8140	JLOW Y TAG	
00060'163000		ADD	3,0	JSET IN CHAR	
00061 040466		STA	Ø, TPTTME		
00062'006457		JSR	echouz	SHIP LOW Y	
00063 000147		TPTTMP		Secondary and second second	
00064'020460		LDA	Ø,TPTX	GET X VALUE	
00065'101112		MOVL#	0,0,SZC		
00066'102400		SUB	0.0		
00067 034450		LDA	3,D1023		
00070'162513		SUBL#	3,0,SNC		
00071 161000		MOV	3,0		
00072'040452		STA	Ø,TPTX	Act Partico In	
00073'101120		MOVEL	0,0	JAND DO LIKE Y	
00074'101120		MOVEL	0.0		
00075101120		MOVEL	0.0	-ISHAP - CAR	
00076'101300		MOVS	0.0	JHI X 5	
00077 034433		LDA	3,8040	JHI X TAG	
00100'163000		ADD	3,0	JADD IN TAG	
00101 040446		STA	Ø,TPTTMP		
00102'006437		JSR	e CHOUZ	SHIP HI X 5	
00103'000147'		TPTTMP			
00104 020440		LDA	Ø,TPTX	JGET X	
00105'034423		LDA	3,8037	GOODIE MASK	
00106'163400		AND	3,0	ILEAVE LOW X 5	
00107'034424		LDA	3, B100	SLOW X TAG	
00110'163000		ADD	3,0	JPUT IN TAG	
00111'040436		STA	Ø,TPTTMP		
00112'006427		JSR	e CHOUZ		
00113'000147'		TPTTMP			
00114'020432		LDA	Ø, TPMOD		
00115'101113		MOVL#	0,0,SNC		
00116'000404		JMP	TPTEXT		
00117'102400		SUB	0,0		
00120'040426		STA	Ø,TPMOD		
00121'000713		JMP	TPTNRM		
00122.050450	TPTEXT:	LDA		JRESTORE ACO	
00123'034420		LDA		JCALL ADDRESS	
00124'001401		JMP	1,3	JEXIT AT CALL+1	
001251000032	SUB00.	032			

00124'001401		JMP
00125'000032	SUBOQ:	032
00126'000033	ESC:	033
00127'000035	GS:	035
00130'000037	US:	037
00131'000020	B020:	020
000130'	BØ37=US	
00132'000040	B040:	040
00133'000100	B100:	100

		1.40			L-50
00134'000140	B140:	140			
00135'000003	DØØ3:	003			
00136'001414	D780:	1414			
00137'001777	D1023:	1777			
00140'000000'	CHINP:	CHIN			
00141'000007'		CHOUT			
00142'000000	TPTACØ:	Ø			
00143'000000	TPTADD:	0			
00144'000000	TPTX:	0			
00145'000000	TPTY:	0			
00146'000000	TPMOD:	Ø			
00147 000000	TPTTMP:	Ø	-	ALLA ALLA ALLA	
00150'040772	CURSIS:	STA		JISAVE ACO	
00151 054772		STA		SAVE CALL ADDRESS	
00152'006767		JSR	echouz	; SET TO ALPHA	
00153'000130'		US			
00154'006765		JSR	echouz	JTURN ON CURSER	
00155'000126'		ESC	-		
00156'006763		JSR	echouz		
00157'000125'		SUBOQ	COULTUD	ACTT CUAD	
00160'006760		JSR	eCHINP	JGET CHAR	
00161'000144'		TPTX		APPE LOOP OCUNTED	
00162'020753		LDA	0,0003	JGET LOOP COUNTER	
00163'040764		STA	Ø, TPTTMF		
00164'020760		LDA	Ø,TPTX	JGET CHAR	
00165'000421	CUDI D.	JMP	CURPS	JSTORE CHAR JGET HI COORD	
00166'006752 00167'000144'	CURLP:	JSR TPTX	BCHTML	JEET HI COORD	
00170'006750		JSR	eCHINP	JGET LOW COORD	
00171'000145'		TPTY	echinr	JULI LOW GOUND	
00172'034736		LDA	3,8037	INASK	
00173'020752		LDA	Ø,TPTY	ILOW COORD	
00174'163400		AND	3,0	MASK OFF GARBAGE	
00175'040750		STA	Ø,TPTY	SAVE FOR LATER	
00176'020746		LDA	Ø,TPTX	;HI COORD	
00177'163400		AND	3,0	IMASK OFF	
00200'101300		MOVS	0.0	ISWAP	
00201'101220		MOVER	0.0		
00202'101220		MOVER	0.0		
00203'101220		MOVER	0,0		
00204'034741		LDA		ILOW COORD	
00205'163000		ADD	3,0	JADD IN LOW COORD	
00206'034735	CURPS:	LDA	3, TPTADE	JCALL ADDRESS	
00207 043400		STA	0,00,3	STORE VALUE	
00210'175400		INC	3,3	ADJUST ADDRESS	
00211'054732		STA	3.TPTADE	SAVE UPDATED ADD	
00212'014735		DSE	TPTTMP	JCHECK FOR DONE	
00213'000753		JMP	CURLP	JLDOP IF NOT	
00214'020726		LDA	Ø,TPTACE	BIRESTORE ACØ	
00215'001400		JMP	0,3	J RETURN	
		.END			

PONT .TITL ROUTINE TO RETURN GLOBAL CO-ORDINATES JOF POINT NP, BLOCK NB. ; INPUT: AC1 = FOINT # NP AC2 = POINTER TO START 3 OF DATA, BLOCK NB. 3 3 ;OUTPUT:ACØ = X CO-ORDINATE AC1 = Y CO-ORDINATE 3 AC2 IS PRESERVED. 3 3 JENTRIES: JSR 0.PON1 . FOR NORMAL ENTRY 3 3 JSR @.PON2 , IF PREVIOUS CALL WAS 3 FOR THIS BLOCK (AC2 3 NOT NEEDED). ; ; .ENT . PON1, . PON2 •EXTD .BSIZ . ZREL 00000-000000' . PON1: PONT1 . PON2: 00001-000170 PONT2 .NREL 00000 054544 PONT1: STA 3,5V3 JIST WORD 00001 '021000 LDA 0,0,2 3,LBIT 00002'034545 LDA JAC3=LONG BLOCK INDICATOR 00003 117400 AND 0,3 00004 054555 STA 3, IND3 ISIN FLAG IN BIT Ø 00005'040547 STA 0,SINF MOVL 0.0 00006 101100 ; COS FLAG IN BIT Ø Ø,COSF 00007 040546 STA **JX CENTROID** LDA 0,1,2 00010'021001 00011 040537 STA Ø,XC 00012 021003 0,3,2 JY CENTROID LDA 00013'040536 STA Ø,YC 00014 021011 LDA 0,11,2 ; SIN STA Ø,SIN 00015 040535 ; COS 0,10,2 00016'021010 LDA 00017 040534 Ø,COS STA BLOCK NB, DATA START 2, SV2 00020 050523 STA 0, BSIZ ; START OF POINT DATA 00021'0200015 ENTO: LDA JPOINTER TO START OF 00022 113000 ADD 0,2 3,3,SER JPOINT LIST 00023 175004 MOV LONG BLOCK 00024 000536 JMP LONG INP*2 FOR SHORT BLOCK 00025 127000 ADD 1,1 (POINT NP) 00026'133000 ADD 1,2 Ø, MASKR 3000000011111111 00027 020516 LDA ; (XR:YR) 00030 025000 LDA 1,0,2 ; (YR:XR) 00031 135300 MOVS 1,3 FRIGHT 8 BITS XR IN AC3 0,3 00032'117400 AND J " " " YR " AC1 00033'107400 AND 0,1 IMASK TO DETECT NEGATIVE 00034 030512 2,0200 LDA 2,1,SZR 00035'147414 AND# IMAKE PROPER NEGATIVE 0,1 00036'106000 ADC 2,3,SZR 00037 157414 AND# J (ALL 16 BITS OK) 0,3 00040'116000 ADC JXR IN AC3, YR IN AC1 1,YR 00041 044515 DOG: STA 2,SIN LDA 00042 030510 0.0 00043'102440 SUBO

00044'125112	MOVL#	1,1,SEC	J-VE YR?
00045'124440	NEGO	1.1	JYES. ABS(YR). SET CARRY
00046'073301	MUL		JYR*SIN IN ACO
00047 125112	MOVL#	1.1.520	ROUNDED ARITHMETIC
00050'101400	INC	0,0	
00051 101002	MOV		RESTORE SIGN
00052 100400	NEG	0.0	
00053'024501	LDA	1,SINF	
00054'125102	MOVL	1,1,SEC	
00055'100400	NEG	0,0	J-VE SIN
00056'024472	LDA	1.XC	
00057 106400	SUB	0,1	JX=XC-YR*SIN
00060 044500	STA	1.X	and the bill
00061 165000	MOV	3,1	
00062'030471	LDA	2,005	
00063 102440	SUBO	0,0	
00064'125112	MOVL#	1,1,SZC	
	NEGO	111320	SET CARRY IF ACI 40
00065'124440		191	JXR*COS IN ACØ
00066*073301	MUL MOVL#	1,1,SZC	JARTOS IN ACO
00067'125112		0,0	
00070'101400	INC	0,0,SZC	
00071 101002	MOV	0,0,520	
00072'100400	NEG	and the second second	
00073'024462	LDA	1,COSF	MUC DELMON BUCKLARDERD
00074'125102	MOVL	1,1,SEC	LUE COS
00075'100400	NEG	0,0	J-VE COS
00076'024462	LDA	1.X	.V-V.VD+COC
00077'107000	ADD	0,1	;X=X+XR*COS
00100'044460	STA	1.X	JGLOBAL X CO-ORD
00101'165000	MOV	3,1	; XR
00102'030450 00103'102440	LDA	2,SIN	
00104'125112	SUB0	0,0	
00105'124440	MOVL#	1,1,SZC	
	NEGO	1,1	NEL CEL
00106'073301	MUL		IXR*SIN
00107'125112	MOVL#	1,1,SZC	
00110101400	INC	0.0	
001111101002	MOV	0,0,SEC	
00112'100400	NEG	0.0	
00113'024441	LDA	1,SINF	
00114'125102	MOVL	1,1,SZC	
00115'100400	NEG	0.0	
00116'024433	LDA	1,YC	
00117107000	ADD	0,1	;YC=YC+XR*SIN
00120'044437	STA	1,1	
00121'024435	LDA	1.YR	
00122'030431	LDA	2,005	
00123'102440	SUBO	0.0	
00124'125112	MOVL#	1,1,SZC	

00125'124440 00126'073301 00127'125112 00130'101400 00131'101002 00132'100400 00133'024422 00134'125102 00135'100400 00136'024421 00137'107000 NEGO MUL MOVL# INC MOV NEG LDA MOVL NEG LDA ADD 1,1,SZC 0,0 0,0,SZC 0,0 1,COSF 1,1,SZC 0,0 1,Y 0,1

IY=Y+YR*COS

00140'020420		LDA	0,X	JOUTPUT	:	XC IN ACØ
00141'030402		LDA	2,5V2	3		YC IN AC1
00142'002402		JMP	esv3	;	ACS	RESTORED
00143'000000	SV2:	Ø				
00144'0000000	SV3:	ø				
00145'000377	MASKR:	377				
00146'000200		200				
00147 020000		20000				
	LBIT:	and the second				
00150'000000	XC:	0				
00151.000000	YC:	0				
00152'000000	SIN:	0				
00153'000000		Ø				
00154'000000		Ø				
00155'000000	COSF:	Ø				
00156'000000	YR:	Ø				
00157 000000	Y:	Ø				
00160.000000	X:	Ø				
00161 0000000	IND3:	0				
00162'135120	LONG:	MOVEL	1,3	INP*3 F	OR LONG	BLOCK
00163'167000		ADD	3,1			
00164'133000		ADD	1,2	POINTE	R TO POI	NT NP (XR)
00165'035000		LDA		XR IN		
00166'025001		LDA	1,1,2	YR IN		
00167 000652		JMP	DOG			
00101 000032	FNTRY			OCK WAS	ADDRESSE	D ON THE LAST
		FOINT IF	INIS DL	UCA MAD	ADDITESSE	D ON THE CAST
	JCALL.	CTA	2 542			
00170'054754	PONT2:	STA	3,5V3			
00171'034770		LDA	3, IND3			
00172'030751		LDA	2,5V2			
00173'000626		JMP	ENTO			
		.END				

PATRY UNPACT CONFE - A TOATA / 126 TRIANS

		.TITL	HITS	
		.ENT	.HITS	
	;			
	ITO SCAN	ALL SI	DES FOR H	IT ON POINT (X,Y)
	3			
	1	JSR e.H	ITS	
	;	X		
	3	Y		
		-HIT RE		
				OCK POINTER
	: 11	AC2.	EDGE # IN	ACI AND BLOCK # IN ACO)
	3 (X)	Y) WILL	BE OVERW	RITTEN WITH THE COORDS
	J OF	THE CEN	TRE OF TH	E LINE THAT WAS HIT
	J ACC	B WILL C	ONTAIN RE	-ENTRY ADDRESS FOR CONTINUED
	I SCA	AN, WITH	RETURN T	O ORIGINAL CALLING ADDRESS.
				TO C(AC3)+1, AC3 WILL BE
	J TAI	KEN AS T	HE NEW CA	ALLING ADDRESS. (GET IT?)
	1			021401000000000000000000000000000000000
		•EXTD	and a second	• M3, M4, M5, M6, M7, MSKR
		•EXTD		PON2, PRN1, EMPT, PSIZ, LENG
		•EXTD	• IACC • • •	PLTS, ALPH
		• ZREL		
00000-000000.	.HITS:	HITS		
		.NREL		
00000'054424	HITS:	STA	3,HIT3	
00001'023400		LDA	00,0,3	
00002'040521		STA	Ø,X	
00003'023401		LDA	0,Y	
00004'040520 00005'0340015		LDA	3. • M1	
00006'102400		SUB	0,0	
00007'040416		STA	Ø,NBB	
00007 040410	BLOCK S	SCAN	U JIVOO	
00010'054416	BEGIN:	STA	3.HOLD	
00011'031400	beorn.	LDA	2,0,3	
00012'151005		MOV	2,2, SNR	
00013'000407		JMP	BAD	INO MORE BLOCKS. EXIT!
00014'024411		LDA	1.NBB	The next presence and the
00015'004412		JSR	SING	GO TO SIDE-SCAN ROUTINE
00016'010407		ISE	NBB	
00017 034407		LDA	3,HOLD	
00020'175400		INC	3,3	
00021 000767		JMP	BEGIN	
00022 034402	BAD:	LDA	3,HIT3	
00023'001402		JMP	2,3	JNO-HIT RETURN
00024'000000	HIT3:	Ø		
00025 000000	NBB:	Ø		
00026.000000	HOLD:	Ø		
	3			
	JINPUT:	ACI - B		
	;	AC2 - P	OINTER TO) START OF DATA, BLOCK NB

00027 054455	SING:	STA
00030'044470		STA
00031'021014		LDA
00032'101005		MOV
00033'002451		JMP
00034'021000		LDA
00035'0240105		LDA
00036'107400		AND

3,SIN3 1,NB 0,14,2 0,0,SNR eSIN3 JZERO AREA. EXIT! 0,0,2 JCONTROL WORD 1,.MSKR 0,1 JNO. OF POINTS

APPROXIMATION.

80037 044446		STA	1.NPNIS	IPOINT COUNTER
00040*126400		SUB	1 + 1	
00041 '044460		STA	1,NP	
00042'006016S		JSR	0.LENG	JGET LENGTH L THIS SIDE
89943'848457		STA	Ø.L	
00044'0068115		JSR	e.PON1	JGET GLOBAL CO-ORDS
00045'040441		STA	0, ×0	
00046'044441		STA	1, 10	
00047 '040444		STA	Ø,XA	
00050'044444		STA	1.YA	
00051 000417		JMP	DOWN	
00052'0060165	BACK:	JSR	8.LENG	JGET LENGTH L
00053'040435		STA	0,L1	JLENGTH L, SIDE NP
20054'0060115		JSR	e.PONI	
00055'040434		STA	Ø,XB	
00056'044434		STA	1,YB	
00057 050423		STA	2,AC2	
00060'034446		JSR	PUSH	JSEARCH FOR CONTACTS
00061 '030421		LDA	2,AC2	
00062'020427		LDA	Ø,XB	INEW BECOMES OLD
00063'040430		STA	C.XA	
00064'020426		LDA	Ø,YB	
00065'040427		STA	Ø,YA	
00066'020422		LDA	0,L1	
00067'040433		STA	Ø,L	
00070'010431	DOWN:	ISE	NP	
00071'024430		LDA	1 . NP	
00072'014413		DSZ	NPNTS	JUMP OUT IF DONE
00073'000757		JMP	BACK	
00074'020412		LDA	0,X0	JLAST LINE
00075 040414		STA	Ø,XB	
00076'020411		LDA	0,40	
00077 040413		STA	Ø,YB	
00100'004426		JSR	PUSH	SEARCH FOR CONTACTS
00101'002403		JMP	esin3	JEXIT
00102.000000	AC2:	Ø		
00103.050000	LBIT:	20000		
00104.000000	SIN3:	0		
00105'000000	NPNTS:	0		
00105'000000	XQ:	0		
00107'000000	YØ:	Ø		
00110.000000	L1:	Ø		
00111.000000	XB:	U		
00115.000000	YB:	0		
00113.000000	XA:	6		
00114'000000	YA:	Ø		
00115'000000	COS:	Ø		
00116,000000	SIN:	Ø		
00117 000000	COSF:	0		
00120.000000	NB:	0		
001011000000	ND+	0		

0 00151,000000 NP: 0 00155.000000 L: 0 X: 00123'000000 00124'000000 Y: 0 0 00125'000000 SINF: 3,SVP3 00126'054541 PUSH: STA JTO GET LOCAL COS AND SIN OF THIS EDGE Part at a tract LDA 0,XB 00127 020762 1 . XA LDA 00130'024763 AX-861 011 SUB 00131'122400

	 -		

00132 040765		STA	Ø,COSF	COS SIGN FLAG
00133101112		MOVL#	0,0,SZC	
00134'100400		NEG	0,0	
00135'030765		LDA		LENGTH OF EDGE
00136'126400		SUB	1,1	
00137'142513		SUBL#	BIT OF	;XD>=L?
00140'124001		COM		JSET AC1 TO 1111
00141 073101		DIV	1717501	VOLT HOT TO TITTE
00142'101112		MOVL#	0.0.570	ROUND UP IF NECESSARY
00143'125400		INC	1,1	THOUGH OF IF NECESSANI
00144'044751		STA	1,COS	
00145'020745		LDA	Ø,YB	
00146'024746		LDA	1,YA	
00147 122400				·VD_VA
00150'040755		SUB	1,0	
				ISIN SIGN FLAG
00151'101112		MOVL#	0,0,SEC	J-VE:
00152'100400		NEG	0.0	
00153'126400		SUB	1,1	
00154'142513		SUBL#		;YD>=L?
00155'124001		COM	1,1,SKP	TES
00156'073101		DIV		
00157'101112		MOVL#	0,0,SEC	-
00160'125400		INC		ROUND UP
00161'044735		STA	1.SIN	
	. CET TO	ANGEOGUE	-	
			D CO-ORDS	
	ICOMPUT		Contraction of the second	YG*SIN(A)
	1	11-10	S+CUSIAI-	-XG*SIN(A)
001/01000741		1.0.4	Ø, X	CET COOPDS OF POINT
00162'020741		LDA	1.Y	JUNDER CONSIDERATION
00163'024741				JUNDER CONSIDERATION
00164'034727		LDA	3,XA	
00165'162400		SUB	3,0	PEL TO FORE STADT
00166'040477		STA	Ø,XG	REL. TO EDGE START
00167'034725		LDA	3.YA	
00170'166400		SUB	3,1	
00171'044475		STA	1,YG	ALAGAL TRANSFORMER V
00172'004477		JSR	YTGET	;LOCAL, TRANSFORMED Y
	;	-		
00173'175112		MOVL#	3,3,SEC	
00174'174400		NEG	3,3	JABS YT
00175'0240175		LDA	1. IACC	
00176'166423		SUBZ		CHECK FOR NORMAL DIST.
00177'002470	YATT I	JMP	esvp3	INOT NEAR; EXIT!
00000100000	;	1.04	0.000	NOW FOR WE DESIGN THE THE
00200'030716		LDA	2,SIN	INOW FOR XT
00201 024465		LDA	1,YG	
00202'102440		SUBO	0.0	
00203'125112		MOVL#	Ist, SEC	JSET CARRY IF NEG

00204'124440 00205'073301 00206'125112 00207'101400 00210'101002 00211'100400 00212'024713 00213'125102 00214'100400 00215'115000 00216'024447

NEGO MUL MOVL# INC MOV NEG LDA MOVL NEG MOV LDA 1,1 JAND MAKE AC1 +VE
1,1,SEC
0,0 JROUND UP
0,0,SEC JCARRY?
0,0 JRESTORE SIGN
1,SINF
1,1,SEC JSIGN OF SIN
0,0
0,3 JSHUNT INTO AC3
1,XG

00217 030676		LDA	2,005		
		SUBO	0.0		
00220'102440		MOVL#	1,1,SEC		
00221'125112		NEGO	1,1		
00222'124440			1 1 1		
00223 073301		MUL	1 1 570		
00224'125112		MOVL#	1,1,SZC		
00225'101400		INC	0.0		
00559,101005		MOV	0,0,SEC		
00227 100400		NEG	0,0		
00230'024667		LDA	1,COSF		
00231'125102		MOVL	1,1,SEC		
00232'100400		NEG	0,0		
00233'117000		ADD	0,3	; ADD	TO PREVIOUS RESULT
	;LOCAL,	TRANSFO	RMED X NO	W IN	AC3
	3				
00234'024666		LDA	LIL		
00235'0200175		LDA	Ø. IACC		
00236'106400		SUB	0,1	1L-5	
00237'166433		SUBZ#	3,1,SNC		
00240'002427		JMP	eSVP3	;OFF	THE END
00241 116433		SUBZ#	0,3,SNC		
00242'002425		JMP	eSVP3	;DITT	ro
	JWE HAVE	E A HIT!			
00243 036425		LDA	3, eHIT3P	2	
00244 020647		LDA	Ø,XA		
00245'024644		LDA	1.XB		
00246'123220		ADDER	1,0		
00247 043400		STA		ISTOR	RE X MID-POINT
00250 020644		LDA	Ø,YA		
00251'024641		LDA	1,YB		
00252'123220		ADDER	1.0		
00253'043401		STA		ISTOR	RE Y MID-POINT
00254'024645		LDA	LINP		
00255'152520		SUBZL	2,2		
00256'146400		SUBLE	2,1		
00257 030623		LDA	2,AC2		
00257 030825					
		LDA	Ø,NB		
00261'005403		JSR	0740073	JHIT	
00262 002405		JMP			RY ON SCAN
00263 056405		STA			RETURN ADDRESS
00264'002403		JMP	esvp3	I CARE	RY ON
00265 000000	XG:	Ø			
00562.000000	YG:	Ø			
00267 000000	SVP3:	Ø			
00270'000024'	HIT3R:	HIT3			
	1				
		CULATE Y			
	J INPUT	YG IN	AC1		
00271 054435	YTGET:	STA	3,YTSAV		
00272 030623		LDA	2,005		
		C 1 1	-		

00273'102440 00274'125112 00275'124440 00276'073301 00277'125112 00300'101400 00301'101002 00302'100400 00303'024614 00304'125102 SUBO MOVL# NEGO MUL MOVL# INC NEG LDA MOVL

1,1,SEC 1,1 1,1,SEC 0,0 0,0,SEC 0,0 1,COSF 1,1,SEC

0.0

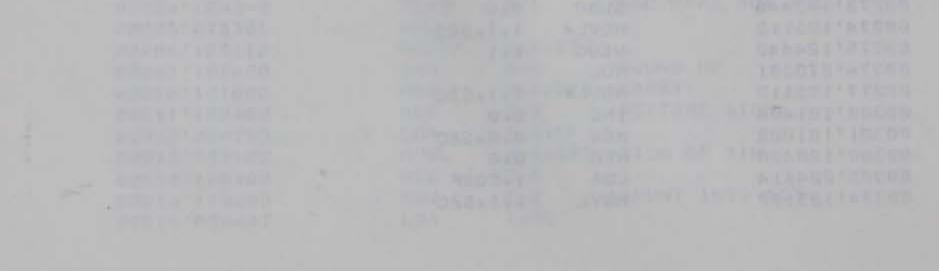
- -NEG 0,0 00305'100400 MOV 0,3 PARTIAL SUM IN AC3 00306'115000 1,XG LDA 00307 024756 2,SIN LDA 00310 030606 SUBO 00311'102440 0,0 1,1,520 MOVL# 00312'125112 NEGO 1,1 00313'124440 MUL 00314'073301 1,1,SZC MOVL# 00315 125112 INC 0,3 00316'101400 MOV 0,0,SEC 00317'101002 0.0 NEG 00320'100400 1,SINF LDA 00321 024604 MOVL 1,1,SEC 00322'125102 0.0 NEG 00323'100400 SUBTRACT FROM PREVIOUS RESULT 0,3 SUB 00324'116400 **eytsav** JMP 00325 002401 YTSAV: 00326'000000 Ø

. END

ANTE VIETTA ANTE IN ALL ANTE VIETTI TA MALL

C-58

CONTAG LECTOR



-					
			.TITL	TAPE	The course of the second second
			. ENT		LNC, .RLNC, .WLNC
			.ENT	.READ.	
			.EXTD	•MEM. •M	1M7
			.ZREL		
	00000-000075'	.OVL:	OVLAY		
	00001-000137'	.CLNC:	CLINC		
	00002-000142*	.RLNC:	RLINC		
	00003-000145*	.WLNC:	WLINC		
	00004-000004'	.READ:	RDP3		
	00005-000000'	.WRIT:	KRTP3		
			•NREL		
		ITUIS D	NITTNE A	LLOLS TH	LISED TO CAUE EN ES
					E USER TO SAVE FILES WRITES (OR READS)
					TAPE (UNIT #1,BLK#150)
					ADS) THE LINKED FIELDS
		A search of a standing of the		BLK#151)	
	00000 054466	WRTP3:	STA	3,RSAVE	
	00001 176400		SUB	3.3	
	00002 054465		STA	and the second se	JSET TO Ø FOR WRITE
	00003 000404		JMP	BEG	JEL TO B FOR WALLE
	00004 054462	RDP3:	STA	3, RSAVE	
	00005 176520	NUFS.	SUBEL	313	
	00006'054461		STA	The second second second	JSET TO 1 FOR READ
	00007 020527	BEG:	LDA	Ø,DRIVE	JOLI TO I FOR READ
	00010'062074	020.	DOB	Ø.LINC	
	00011'020454		LDA	ØFBLK	
	00012'126520		SUBEL	1.1	JONE BLK FOR PAGE ZERO
	00013'152400		SUB	2,2	ISTART AT LCTN Ø
	00014'034453		LDA	3.FLAGE	
	00015 175004		MOV	3.3.SZR	
	00016'000402		JMP	READF	
	00017'000406		JMP	WRITE	
	00020'006002-	READE :	JSR	.RLNC	
	00021'125005		MOV	1,1,SNR	
	00022'000410		JMP	NXT1	
	00023'063077		HALT		
	00024'000763		JMP	BEG	
	00025'006003-	WRITE:	JSR	e-WLNC	
	00026'125005		MOV	1.1. SNR	
	00027 000403		JMP	NXT1	
	00030'063077		HALT		
	00031 000756		JMP	BEG	
	00032'020504	NXT1:	LDA	Ø.DRIVE	
	00033'062074		DOB	Ø.LINC	
	00034'0240035		LDA	1 M7	JDETERMINE LENGTH OF
	00035'0300025		LDA	2MI	JLINKED FIELDS IN USE
	00036'146400		SUB	2,1	
	00037 030425		LDA	2,0400	

00040'102400 00041'073101 00042'020423 00043'101400 00044'125400 00045'0300025 00046'034421 00047'175004 00050'000402 00051'000406 SUB DIV LDA INC LDA LDA MOV JMP JMP

Ø,FBLK Ø,Ø ;START AT F 1,1 ;ADD AN EXT 2,.M1 ;START @ LI 3,FLAGF 3,3,SER READG WRITG

0.0

JSTART AT FBLK+1 JADD AN EXTRA BLOCK JSTART & LINKED LISTS

				C-60
00052.0006005-	READG:	JSR	e.RLNC	0.00
00053'125005		MOV	1,1, SNR	
00054'002412		JMP	e RSAVE	
00055'063077		HALT		
00056'000754		JMP	NXT1	
00057 006003-	WRITG:	JSR	e.WLNC	
00060'125005		MOV	1,1, SNR	
00061 002405		JMP	ERSAVE	
00062'063077		HALT		
00063 000747		JMP	NXT1	
00064.000400	C400:	400		
00065'000150	FBLK:	150		
00066'000000	RSAVE:	Ø		
00067'000000	FLAGF:	Ø		
	3			
				RLAY NUMBER 1
	57 AL 1005510 1000	and the second second		FIRST TRANSFERING
	JITSELF	TO A SAF	TE PLACE	IN HIGH CORE.
00070 000000	NUB:	Ø		INO NEED TO TRANSFER P-3 R&W
00071 000002	TWO:	2		ROUTINES SO START AT NUB
00072'000003	THREE:	3		
00073 000070	FIRST:	NUB		
00074'000326'	LAST:	CB		
00075 020441	OVLAY:	LDA	Ø, DRIVE	
00076'062074	OVLAT.	DOB	ØLINC	
00077'0340015		LDA	3. MEM	HIGHEST MEMORY LCTN
00100'030773		LDA	2.FIRST	MICHEST MENONI LOIN
00101 020773		LDA	Ø,LAST	
00102'142400		SUB	2,0	SENUMBER OF WORDS TO BE MOVED
00103'101400		INC	0.0	
00104'116400		SUB	0,3	INEW ADDRESS
00105'100400		NEG	0.0	10.2 CEASTRYAINS
00106'025000	ROUND:	LDA	1,0,2	
00107 '045400		STA	1,0,3	
00110'101405		INC	0,0, SNR	
001111000404		JMP	OUT	
00112'151400		INC	2,2	
00113'175400		INC	3,3	
00114'000772		JMP	ROUND	
00115'156400	OUT:	SUB	2,3	J=DISTANCE MOVED
00116'030403		LDA	2, SHIFT	
00117'157000		ADD	2,3	
00120'001400		JMP	0.3	J GO TO HI-CORE COPY
00121'000122'	SHIFT:	• + 1		
00122'020412		LDA	Ø,BLK1	
00123'024412		LDA	1,NBLK1	
00124'152400		SUB	2,2	
00125'004415		JSR	RLINC	
00126'125005		MOV	1,1, SNR	
00127'000377		JMP	377	FORTRAN START ADDRESS
00130'063077		HALT		JLINC ERROR
00131 020405		LDA	0.DRIVE	JIRY AGAIN (PRESS CONTINUE)

 00131.020405
 LDA

 00132.062074
 D0B

 00133.000767
 JMP

 00134.000350
 BLK1:

 00135.000055
 NBLK1:

 00136.000001
 DRIVE:

 1

LDA Ø,DRIVE JTRY AGAIN (PRESS CONTINUE) DOB Ø,LINC JMP SHIFT+1 BLK1: 350 NBLK1: 55 DRIVE: 1 JNOW FOLLOWS THE STANDARD LINCTAPE JUTILITIES...

•						
		; INPUT:	ACØ	=FIRST BLO	CK	
		3	AC1	=NUMBER OF	BLOCKS	
		;	AC2	=FIRST COR	E ADDRESS	5
		;				
		JOUTPUT:	AC1	=ERROR COD	E	
		;				
	00137'054430	CLINC:	STA	3, SAC3		
	00140'152400		SUB	2,2		
	00141'000417		JMP	CHKE		
	00142'054425	RLINC:	STA	3, SAC3		
	00143'034430		LDA	3,D2R		
	00144'000415		JMP	READZ		
	00145'054422	WLINC:	STA	3, SAC3		
	00146'034423		LDA	3.D1W		
	00147'054510		STA	3,D1XX		
	00150'044501		STA	1,D2XX		
	00151'050417		STA	2, SAC2		
	00152'004423		JSR	DO		
	00153'024476	RAW:	LDA	1,D2XX		
	00154'122400		SUB	1,0		
	00155'030413		LDA	2,SAC2		
	00156'151113		MOVL#	2,2, SNC		
	00157'150000		COM	2,2		
	00160'034473	CHKZ:	LDA	3,D2C		
	00161 054470	READZ:	STA	3.D2XX		
	00162'034410	NLADE.	LDA	3.DIRC		
	00163'054474		STA	3.DIXX		
	00164'004411		JSR	DO		
	00165'060274	EXIT:	NIOC	LINC		
	00166'002401	Lerie .	JMP	eSAC3		
	00167'000000	SAC3:	Ø	CONCO		
	00170'000000	SAC2:	ø			
	00171 021000	D1W:	LDA	0,0,2		
	00172'000750	DIRC:	JMP	READ-DIX	XAL	
	00173'132512	D2R:	SUBL#	1,2,SEC		
	00174'000000	RETU:	0			
	00175'054777	DO:	STA	3,RETU		
	00176'075474		DIB	3.LINC		
	00177'175112		MOVL#	3,3,52C		
	00200'000446		JMP	E4		
	00201'151113		MOVL#	2,2, SNC		
	00202'000410		JMP	FINDF		
	00203'150000		COM	2,2		
	00204'176400	FINDR:	SUB	3,3		
	00205'162000	, and the	ADC	3,0		
	00206'060374		NIOP	LINC		
	00207 004467		JSR	GETBL		
	00210'101401	FINDN:	INC	0,0,SKP		
	00211'000776		JMP	2		
	00212'060174	FINDF:	NIOS	LINC		
	000121000114	r inpr .	ISD	CETRI		

00213'004463 00214'000777 00215'175224 00216'000766 00217'125005 00220'002754 00221'166000 00222'040474 00223'044474 00224'024476

JSR JMP MOVZR JMP FOUND: MOV JMP ADC STA

STA

LDA

GETBL •-1 R 3,3,SZR FINDR 1,1,SNR eRETU 3,1 0,TEMP1 1,TEMP2 1,SIZE

				C-62
00225 1 47000		ADD	2,1	
00226'000431		JMP	DIXX	
00227 063674	READ:	SKPDN	LINC	
00230'000777		JMP	• - 1	
00231 063474		SKPBN	LINC	
00232'000416		JMP	RDAT	
00233'060474	RCHK:	DIA	Ø,LINC	
00234'116405		SUB	0,3, SNR	
00235'000434		JMP	SCHK	
00236'024465	E1:	LDA	1.01	
00237 000403		JMP	•+3	
00240'034462	E2:	LDA	3,SIZE	
00241 024463		LDA	1,02	
00242'020454		LDA	Ø,TEMP1	
00243 000722		JMP	EXIT	
00244 024461	E3:	LDA	1,04	
00245 000720		JMP	EXIT	
00246 024460	E4:	LDA	1,08	
00247 000716		JMP	EXIT	
00250'060474	RDAT:	DIA	Ø,LINC	
00251 132512	DSXX:	SUBL#	1,2,SZC	
00252'041000		STA	0,0,2	
00253'000402	DSC:	JMP	•+2	
00254 061074	WDAT:	DOA	Ø,LINC	
00255'117000	BLOOP:	ADD	0,3	
00256 151400		INC	2,2	
00257 021000	DIXX:	LDA	0,0,2	
00260 063074		DOC	Ø,LINC	
00261 063674		SKPDN	LINC	
00262'000777		JMP	·-1 0013	
00263 063474		SKPBN	LINC	
00264'000770		JMP	WDAT	
00265 075074	WCHK:	DOA	3.LINC	
00266'075474		DIB	3,LINC	
00267 175004		MOV	3,3,SZR	
00270 000756		JMP	E4	
00271'132414	SCHK:	SUB#	1,2,SZR	
00272'000746	NEVT-	JMP	E2	
00273'020423	NEXT:	LDA	Ø,TEMPI	
00274'024423		LDA	1, TEMP2	
00275'000713	CETRI .	JMP	FINDN	
00276 054420	GETBL:	STA	3, TEMPI 3, MLIM	
00277'034421		LDA SUBZ#	3,0,SZC	
00300'162432		JMP	WAIT	
00301'000405 00302'034417		LDA	3.PLIM	
00303'162032		ADCZ#	3,0,SZC	
00304'000740		JMP	E3	
00305 074474		DIA	3,LINC	
00306 063474	WAIT:	SKPBN	LINC	
00307 000777		JMP	WAIT	
00310'063774		SKPDZ	LINC	
		AUT 1 1 1 1 1 1 1 1 1 1 1 1 1 1 1 1 1 1 1		

00310 003/14 00311 000774 00312'074474 00313'116543 00314'010402 00315'002401 00316'000000 TEMP1: 00317'000000 TEMP2: 00320'177770 MLIM:

SKIDE P 1110 WAIT-1 JMP DIA 3.LINC 0,3, SNC SUBOL ISZ TEMPI JMP **ØTEMP1** 0 Ø 177770

00321'000620	PLIM:	620
00322'000400	SIZE:	400
00323'000001	C1:	Interference and a state of the second state of the
00324'000002	C2:	2
00325'000004	C4:	4
00326'000010	C8:	10
		• END

FIRST TRATEGORY INTO A THE ART OF THE ACT OF THE ACT.

-				
			.TITL	UTIL
		J SEVERAL	UTILITY	PROGRAMS
			.ENT	.HITC IACC PRN1 PAGE LENG SCAL
			.ENT	.VFAC IPRN PRN2 MESS ALPH TYP
			.ENT	.AXIS, GETT, DBIN, CHEK, WORD, DB0
			.EXTD	.M1, .DISS, .LPAP, .MSKR, .PLTS
			.ZREL	
	00000-000005	.IACC:	5	
	00001-0000000	.HITC:	HITC	
	00002-000052*	.PRN1:	PRN1	
	00003-000270	.PRN2:	PRN2	
	00004-000164'	. IPRN:	TART	
			MESS	
	00005-000331'	.MESS:	1200 1200 1200	
	00006-000655'	.WORD:	WORD	
	00007-000062'	.ALPH:	ALPHA	
	00010-000067'	.PAGE:	PAGE	
	00011-000101'	·LENG:	LENG	
	00012-000126'	.TYP:	TYPE	
	00013-000151'	.SCAL:	SCAL	
	00014-000421'	.AXIS:	AXIS	
	00015-000560'	•GETT:	GET	
	00016-000572*	.DBIN:	DBIN	
	00017-000570'	•DB0:	DBØ	
	00020-000640'	. CHEK:	CHEK	
	00021-000003	.VFAC:	3	
			•NREL	
		3		
				WHICH BLOCK HAS CENTROID
		ICORRESI	PONDING T	O GIVEN X,Y CO-ORDINATE
		3		
		1	JSR e.HI	
		3	25 SSE	DRESS OF INPUT X)
		3	and a second a second a second as	DRESS OF INPUT Y)
		C.4 20150000		IF NO HIT)
		I (RE		WITH POINTER TO BLOCK
		3	IN AC2 1	F SUCCESSFUL, AND NB IN ACI)
		1		
	00000'023400	HITC:	LDA	0,00,3
	00001 040445		STA	Ø, X
	00002'023401		LDA	0,01,3
	00003 040444		STA	Ø,Y
	00004'054444		STA	3, SVH3
	00005'102400		SUB	0,0
	00006'040443		STA	Ø,NB
	00007 0340015		LDA	3, •M1
	00010'031400	LOOP:	LDA	2,0,3
	00011 151005		MOV	2,2, SNR
	00012 000432		JMP	NOHIT JLAST BLOCK
	00013'021014		LDA	0,14,2
	00014'101005		MOV	0,0,SNR
	000151000404		IND	NEVT SECO ADEA

199-3

00015'000424 00016'021001 00017'024427 00020'122400 00021'101112 00022'100400 00023'024000-00024'106512 00025'000414 00026'021003 JMP LDA LDA SUB MOVL# NEG LDA SUBL# JMP LDA NEXT JEERO AREA 0,1,2 JXC 1,X 1,0 0,0,SEC 0,0 JABS(XC-X) 1,.IACC 0,1,SEC NEXT JNOT THIS BLOCK 0,3,2 JYC

-						
	00027 024420		LDA	1.7		
	00030'122400		SUB	1,0		
	00031'101112		MOVL#	0,0,SEC		
	00032'100400		NEG	0,0	JABS(YC-Y)	
	00033'024000-		LDA	1. IACC		
	00034'106512		SUBL#	0,1,SZC		
	00035'000404		JMP	NEXT		
	00036'034412		LDA	3,SVH3	JMUST BE HIT	
	00037'024412		LDA	1,NB		
	00040'001403		JMP	3,3	JGOOD EXIT	
	00041 175400	NEXT:	INC	3,3	FOOD LAIT	
	00042'010407	NEAT.	ISE	NB		
	00043'000745		JMP	LOOP		
	00044'034404	NOHIT:	LDA	3, SVH3		
	00045 001402	NOATI.	JMP	2,3	JBAD EXIT	
		×.		235	JOHD CALL	
	00046'000000	X:	0			
	00047 000000	Y:	Ø			
	00050'000000	SVH3:	0			
	00051 000000	NB:	Ø			
		1				
				and a survey of the second s	RACTER, WAITING	
		JUNTIL	THE TTY	IS FREE.		
		3				
		1	JSR e.F			
		3	N (N		CHARACTER TO BE	
		1 (1-14)	1.1		(NOT ADDRESS])	
		3	(A	CCUMULAT	ORS ARE SAVED)	
		;	0.000			
	00052'040407	PRN1:	STA	Ø,ACØSV		
	00053'021400		LDA	0,0,3		
	00054'063511	PRH:	SKPBZ	TTO		
	00055.000111		JMP	•-1		
	00056'061111		DOAS	Ø,TTO		
	00057'020402		LDA	Ø,ACØSV		
	00060'001401		JMP	1,3		
	00061 1000000	ACØSV:	Ø			
		3				
		JTO SET	TEKTRON	IX TO AL	PHA MODE	
		3	JSR e.A	LPH		
		3				
	00062'054404	ALPHA:	STA	3,ASAV		
	00063'004767		JSR	PRN 1		
	00064'000037		37			
	00065'002401		JMP	BASAV		
	00066'000000	ASAV:	Ø			
		3				
		JTO ERAS	SE SCREE	N		
		3				
		3	JSR @.F	AGE		
		3				
	00067 054410	PAGE:	STA	3,SVP3		
	00070'004762		JSR	PRN1		
	000011000000		22			

00071 000033 00072'004760 00073'000014 00074'102400 00075'0400035 00076'002401 00077 000000 SVP3: Ø

3

JSR PRN1

33

14

SUB

STA

JMP

0,0 JSUPPRESS HARD-COPY 0. LPAP ILOAD PLOTTING THE CALLER @SVP3

C-65

100-0

;ROUTINE TO RETURN LENGTH, L OF SIDE NP J JSR Q.LENG J INPUT: AC1 - SIDE # (NP) AC2 - POINTER TO BLOCK DATA 3 ; OUTPUT: ACO - LENGTH L 000025 START=25 ; POINT DATA STARTS AT 25RD WORD 000026 SS=START+1 000027 SL=START+2 7777 ; TO REMOVE TYPE # 00100'007777 TMSK: STA 3, SVP3 LENG: 00101 054776 0,0,2 ; CONTROL WORD 00102 021000 LDA 00103'034420 3,LBIT LDA AND# 0,3,SER JLONG BLOCK? 00104 117414 JMP JYES 00105'000407 LONG ;NP*2 00106'135120 MOVEL 1,3 ADD 2,3 00107 157000 LDA Ø,SS,3 JGET L 00110'021426 LDA 3, TMSK 00111'034767 00112'163400 AND 3.0 JMP eSVP3 JEXIT WITH L IN ACØ 00113 002764 LONG: MOVEL 00114'135120 1,3 INP*3 00115'137000 ADD 1,3 00116'157000 ADD 2,3 00117 021427 0, SL, 3 LDA LDA 3,TMSK 00120'034760 00121'163400 AND 3,0 00122 002755 JMP @SVP3 JEXIT 00123'020000 LBIT: 20000 3 ;ROUTINE TO RETURN SURFACE TYPE # JFOR A GIVEN EDGE JSR @.TYP 1 JINPUT: AC2 = DATA POINTER FOR GIVEN BLOCK J = AC1 = EDGE # (NP)JOUTPUT: ACØ = TYPE # AC1 AND AC2 ARE PRESERVED 3 00124'170000 LMSK: 170000 FOR MASKING OUT LENGTH PART TSAV: 00125'000000 Ø 00126'054777 TYPE: STA 3,TSAV 00127 021000 0,0,2 JCONTROL WD LDA LDA 3,LBIT 00130'034773 00131 117414 AND# 0,3,SER 00132'000405 JMP LONGI MOVEL 00133'135120 1,3 00134'157000 ADD 2,3 LDA 00135 021426 0,55,3 NOSE 00136'000405 JMP

			the second second second	
00137 135120	LONG1:	MOVEL	1,3	
00140'137000		ADD	1,3	
00141'157000		ADD	2,3	
00142'021427		LDA	0, SL, 3	
00143'034761	NOSE:	LDA	3,LMSK	
00144'163700		ANDS	3,0	
00145'103120		ADDEL	0.0	
00146'103120		ADDZL	0.0	
00147 101300		MOVS	0.0	
00150'002755		JMP	etsav	

	IVECTOR	SCALING	POLITINE	C-67
00151'030021-		LDA	2. VFAC	
00152'102400	JUNE .	SUB	0,0	
00153'044410		STA	1,AC1	
00154'125112		MOVL#	I.I.SEC	
00155'124400		NEG	1,1	
00156'073101		DIV	0.404	
00157 030404		LDA	2.AC1	
00160'151112		MOVL#	2,2,SZC	
00161 124400		NEG	1,1	
00162'001400		JMP	0,3	
00163'000000	AC1:	Ø		
	3	Sec. Ender		
	IROUTIN		NT A RIGHT-JUSTIFIED	
•			D LENTH, WITH LEADI	NG ZEROS
	JOR WITH	HOUT		
	3			
	3	JSR e.II		
	3 (-)	N N	(VALUE, NOT ADDRESS	;)
	1			
	3		IS FIELD LENGTH (ZE	RUS PRINIED
	3	IF NEGAT		Contractor in the second
	3	THE NUME	BER TO BE PRINTED IS	IN ACØ
	J			
00164'031400	TART:	LDA	2,0,3	
00165'101112		MOVL#	0,0,SZC	
00166'100400		NEG	0.0	
00167 175400		INC	3,3	
00170'054524		STA	3, SAV3	
00171'151112		MOVL#	2,2,SZC	
00172'150401		NEG	2,2,SKP	
00173'126401		SUB	1,1,SKP	
00174'126520		SUBEL	1,1	
00175'044520		STA	1, FLAG JSTORE ZERO	IBLANK FLAG
00176'050520		STA	2, FIELD ; FIELD LENG	TH
00177'034475		LDA	3, TENS	
00200'054517		STA	3, POINT	
00201 034502		LDA	3,HOLD	
00202'054516		STA	3, PPNT	
00203'034507		LDA	3, JOLD	
00204 054414		STA	3.MM	
00205'152400		SUB	2,2	
00206'036511	BIG:	LDA	3, POINT	
00207 010510		ISE	POINT	
00210'175005		MOV	3,3, SNR	
00211'000416		JMP	END	
00212'126400		SUB	1,1	
00213'162422	SMALL:	SUBE	3,0,52C	
00214'125401		INC	1,1,SKP	
00215'163001		ADD	3,0,SKP	
00216'000775		JMP	SMALL	
00217 046501		STA	1, OPPNT	
000001105015		HOUH	I I CALD	

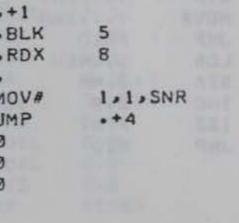
00220'125015 MM: 00221'000404 00222'034471 00223'054775 00224'151400 00225'010473 FRED: 00226'000760

MOV# JMP LDA STA INC ISZ JMP 1,1,SNR FRED 3,JNEW 3,MM 2,2 PPNT BIG

JCOUNT NON-ZERO DIGITS

					0-00
00227 034467	END:	LDA	3.FIELD		
00230'151005		MOV	2,2, SNR		
00231'151400		INC	2,2		
00232'050467		STA	2, SAV2		
00233'156423		SUBZ	2,3, SNC		
00234'000427		JMP	ASTER	FIELD TOO SMALL	
00235'170405		NEG	3,2, SNR		
00236'000410		JMP	DIGIT	INO ZEROS	
00237 024456		LDA	1.FLAG		
		LDA	ØJEERO		
00240'020463		and a strend of the second sec	1,1,SNR		
00241'125005		MOV	Ø, BLANK		
00242'020462		LDA	0. PRN2	SEND OUT LEADING	
00243 006003-		JSR	and the second se		
00244'151404		INC	- The second second	JZEROS OR BLANKS	•
00245'000776	_	JMP	2		
00246'030443	DIGIT:	LDA	2,BOT		
00247 024452		LDA	1, SAV2		
00250'132400		SUB	1,2		
00251 124405		NEG	1,1,SNR		142
00252'002442		JMP	esav3	SNOTHING TO PRINT	
00253 021000	LOOP1:	LDA	0.0.2		
00254'034447		LDA	3,ZERO		
00255'163000		ADD	3.0		
00256'006003-		JSR	e.PRN2	SEND OUT DIGIT	
00257'151400		INC	2,2		
00260'125404		INC	1,1,SZR		
00261 '000772		JMP	LOOPI		
00262'002432		JMP	eSAV3	JEXIT	
00263 020437	ASTER:	LDA	Ø,AST	JSEND OUT ASTERISKS	
00264'006003-	NIT:	JSR	e.PRN2		
00265'014431		DSE	FIELD		
00266'000776		JMP	NIT		
00267 002425		JMP	eSAV3		
- CONTRACTOR	1				
	ROUTINE	TO PRIM	NT OUT SI	INGLE CHARACTER	
	1	JSR e.P			
	JINPUT:	CHARACTI	ER IN ACC	0	
	;				
00270'063511	PRN2:	SKPBZ	TTO		
00271 000777		JMP	1		
00272'061111		DOAS	Ø,TTO		
00273'001400		JMP	0,3		
00010 001400	1				
000012		.RDX	10		
00274'000275'	TENS:	• + 1	12020		
00275'023420	· Liver	10000			
00276'001750		1000			
00277 000144		100			
00300.000015		10			
00301 0000012		1			
00302 000000		ø			
000002 0000000					

00303'000304'	HOLD:	•+1
000005		.BL
800010		• RI
00311'000311'	BOT:	
00312'125015	JOLD:	MON
00313'000404	JNEW:	JMI
00314'000000	SAV3:	Ø
00315'000000	FLAG:	Ø
00316'000000	FIELD:	Ø



 				L-09
00317 '000000	POINT:	Ø		
00320'000000	PPNT:	Ø		
00321 '000000	SAV2:	Ø		
00322'000052	AST:	***		
00323'000060	ZERO:	···Ø		
00324'000040	BLANK:			
	1			
	JTO PRI	NT MESSA	GE ON SCI	REEN AT
	JA SPEC	IFIC LOC	ATION	
	1			
	;	JSR 0.M	ESS	
	1	TEXT	(ADDRESS	S OF TEXT)
	1 (-	X	(X,Y LOO	CATION OF MESSAGE
	;	Y		VALUES, NOT
	1		ADDRES	SESI. NEGATIVE X DRAWS
	1			UNDER TEXT)
	1			The second s
00325'000000	FLAG1:	Ø		
00326'000000	MSAV:	0		
00327 000000	BPNT:	ø		
00330'000000	COUNT:	ø		
00331'021400	MESS:	LDA	0,0,3	
00332'101120	112021	MOVEL	0.0	CREATE BYTE POINTER
00333'040774		STA	Ø, BPNT	
00334'021401		LDA	0,1,3	JX
00335'101112		MOVL#	0,0,SEC	
00336'100401		NEG	0.0. SKP	
00337'126401		SUB	1,1,SKP	
00340'126520		SUBEL	1,1	
00341 044764		STA	1.FLAG1	
00342'025402		LDA	1,2,3	JY
00343'054763		STA	3.MSAV	A DAM & MEN
00344'040451		STA	Ø,XSAV	FREMEMBER X & Y FOR
00345'044451		STA	1.YSAV	JLATER PLOTTING OF LINE
00346'0060055		JSR	8.PLTS	JINITIALISE BEAM
00347 000000		Ø		BEAM OFF
00350'006007-		JSR	e.ALPH	
00351'102400		SUB	0.0	
00352'040756		STA	Ø, COUNT	
00000 040100	ROUTIN			UNTIL ZERO BYTE FOUND
00353 030754	PICK:	LDA	2, BPNT	
00354'010753		ISZ	BPNT	
00355'151220		MOVER	2,2	
00356'021000		LDA	0,0,2	
00357 030004S		LDA	2. MSKR	
00360'101002		MOV	0,0,SEC	
00361 101300		MOVS	0,0	
00362 143405		AND	2,0, SNR	
00363'000404		JMP	RET	
00364 010744		ISZ	COUNT	
00365'006003-		JSR	e.PRN2	JSEND OUT CHARACTER
00000 000000-		USIN	C TI MIL	

00366'000765 JMP PICK 00367 020736 RET: LDA Ø,FLAG1 MOV 0,0, SNR 00370'101005 JMP 00371'000422 PAST ITO PLOT LINE UNDER TEXT 00372'024424 LDA 1,YSAV Ø,GAP LDA 00373'020424 0,1 SUB 00374'106400 1,YSAV STA 00375'044421

					C-70
00376'020417		LDA	Ø,XSAV		
00377 .0060055		JSR	e.PLTS	JFIRST END OF LINE	
00400'000000		Ø			
00401 102400		SUB	0,0		
00402'024416		LDA	1,N14		
00403'030725		LDA	2,COUNT		
00404'073301		MUL			
00405'020410		LDA	Ø,XSAV		
00406 123000		ADD	1,0		
00407 024407		LDA	1.YSAV		
00410'0060055		JSR	P.PLTS	SECOND END	
00411 000001		1			
00412'006007-		JSR	e.ALPH		
00413 034713	PAST:	LDA	3.MSAV		
00414'001403	1	JMP	3,3	JEXIT	
00415 000000	XSAV:	0	0,0		
00416'000000	YSAV:	Ø			
00417 000003	GAP:	3	IGAP BE	TWEEN TEXT AND LINE	
00420'000016	N14:	16		OF ONE LETTER	
00420 000010	1 1	10		or one perfer	
	TO DRA	W A SCAL	F WITH 1	Ø TICK MARKS,	
				, WITH THE	
		ABOVE OR			
	I	ABOVE ON	DELON R		
		JSR P.A	XIS		
			LENGTH	1)	
		(-) X	(STARTI		
				CO-ORD)	
		CALL AR	the set of	ARE VALUES, NOT	
		a state of the second	ESSES)		
	1	noon			
	ITE I H	AS - SIG	N. AXIS	WILL BE PARALLEL	
				ARALLEL TO X AXIS	
	1				
	JIF X H	AS - SIG	N. TICKS	WILL BE BELOW	
		OTHERWIS			
	1				
00421 054521	AXIS:	STA	3.TTSAV	the second second	
00422 021400		LDA	0,0,3		
00423 101112		MOVL#	0,0,SEC	OT TRITINGS	
00424'100401		NEG	0.0. SKP	And Party Arthre	
00425'126401		SUB	1,1,SKP	The second	
00426 126520		SUBEL	1,1		
00427 044517		STA	1.FLOG	JX/Y FLAG	
00430 040505		STA	ØoL		
00431'021401		LDA	0,1,3		
00432'101113		MOVL#	0,0,SNC	NOT THE REAL	
00433'000405		JMP	ABOVE		
00434'100400		NEG	0.0		
00435'024512		LDA	1,TICB		
00436'044455		STA	1,REPL		
00437 000403		JMP	GETY		
001101001F10	ADOUR .	1 0 4	I. TICA		

00440'024510 00441'044452 00442'040474 00443'025402 00443'025402 00445'030470 00445'030470 00445'151220 00447'151220 LDA STA STA LDA STA LDA MOVZR MOVZR

ABOVE:

GETY:

1,TICA 1,REPL 0,XN 1,2,3 1,YN 2,L 2,2 2,2

-							
	00450'151220		MOVER	2,2			C-71
	00451 151220		MOVER	2,2			6-71
	00452'151220		MOVER	2,2			
	00453 050465		STA	2,11			
	00454'147000		ADD	2,1			
	00455'004474		JSR	PLOT			
	00456'000000		Ø	1 201			
	00457 020457		LDA	Ø,XN			
	00460'024457		LDA	1. YN			
	00461 004470		JSR	PLOT			
	00462 000001		JUSIC	FLUI			
			100	0.XN			
	00463'020453		LDA				
	00464'024453		LDA	1. YN			
	00465'030450		LDA	2,L			
	00466'143000		ADD	2,0			
	00467 004462		JSR	PLOT			
	00470'000301		1	~			
	00471'020445		LDA	Ø, XN			
	00472 024445		LDA	1. YN			
	00473'030442		LDA	2,1			
	00474'143000		ADD	2,0			
	00475'030443		LDA	2,11			
	00476'147000		ADD	2,1			
	00477'004452		JSR	PLOT			
	00500'000001		1				
	00501'102400		SUB	0.0			
	00502'024433		LDA	1.1			
	00503'030440		LDA	2,NINE			
	00504'050440		STA	2,TCNT			
	00505'151400		INC	2,2			
	00506'073101		DIV				
	00507 044436		STA	1,DIVIS			
	00510'020430		LDA	Ø,LI			
	00511'101220		MOVER	0,0			
	00512'024425		LDA	1.YN			
	00513'107000	REPL:	ADD	0.1	ITHIS W	ORD CAN BE CH	ANGED
	00514'044425		STA	1.YN1	1		
	00515'024422	TEA:	LDA	1. YN	JTO PLO	T TICKS ON AN	IS
	00516'020420		LDA	Ø,XN			
	00517'030426		LDA	2,DIVIS			
	00520'143000		ADD	2,0			
	00521 040415		STA	Ø,XN			
	00522'004427		JSR	PLOT			
	00523'000000		Ø				
	00524'020412		LDA	Ø,XN			
	00525'024414		LDA	1. YNI			
	00526'004423		JSR	PLOT			
	00527'000001		1				
	00530'014414		DSE	TCNT			
	00531 000764		JMP	TEA			
	00532'006007-		JSR	e.ALPH			
	00533'034407		LDA	3,TTSAV			
	00555 054407		LUA	2 2			

00534'001403 00535'000000 L: 00536'000000 XN: 00537'000000 YN: 00540'000000 L1: 00541'000000 YN1 00542'000000 TTS 00543'000011 NIN

L: XN: YN: L1: YN1: TTSAV: NINE:

JMP 3,3 0 0 0 0 0 0 0 0 0 11

00544'000000	TCNT:	Ø			
00545'000000	DIVIS:	Ø			C-72
00546.000000	FLOG:	Ø			24 14 254
00547'106400	TICB:	SUB	0,1		
00550 107000	TICA:	ADD	0,1		
00551 030775	PLOT:	LDA	2,FLOG		
00552'151005		MOV		X OR Y AXIS?	
00553 000404		JMP	JOE		
00554'111000		MOV	0,2		
00555'121000		MOV	1,0		
00556'145000		MOV	2,1		
00557'0020055	105 .	JMP	e.PLTS		
00331 0020033	302.	Juit	errurs		
	TO CET	A TTY C	HARACTER		
	10 021		0.GETT		
	OUTOUT	JSR .		ca.	
	JUUIPUI	: CHARAC	CTER IN A		
	,	CUDON	***		
00560 063610	GET:	SKPDN	TTI		
00561 000777		JMP	•-1		
00562 060510		DIAS	Ø,TTI		
00563 101300		MOVS	0.0		
00564'101120		MOVEL	0.0		
00565'101220		MOVER	0.0		
00566'101300		MOVS	0.0		
00567 001400		JMP	0,3		
	3				
	3				
	the second second second second			INE CALMOST	
	JIDENTI		ATA GENER	RAL'S)	
	3	JSR e.E			
	JOUTPUT	:	# IN AC	1 1012.48.584.69	
	;				
00570'054443	D80:	STA	3,DBSAV		
00571 000403		JMP	DB1		
00572'054441	DBIN:	STA	3.DBSAV		
00573 006015-		JSR	e.GETT		
00574'126400	DB1:	SUB	1,1	SENTRY WITH FIRST	
00575'044437		STA	1.EC10	CHARACTER IN ACO	
00576'044437		STA	1,EC11		
00577 024437		LDA	1,EC20		
00600*106405		SUB	0,1, SNR		
00601 000405		JMP	EC96		
00602'024435		LDA	1,EC21		
00603'106404		SUB	Ø,1,SER		
00604 000404		JMP	EC98		
00605 010427		ISE	EC10		
00606 006003-	EC96:	JSR	e.PRN2		
00607 006015-	EC97:	JSR	e.GETT		
00610 006003-	EC98:	JSR	e.PRN2		
00511 006020-		JSR	e.CHEK		
00612'000405		JMP	EC95		
00613'024422		LDA	1.ECI1		
00/11/1001111		100	FOFO		

00614'004411 00615'044420 00616'000771 00617'024416 EC95: 00620'125120 00621'014413 00622'125221 00623'124640 JSR STA JMP LDA MOVZL DSZ MOVZR NEGOR EC50 1,EC11 EC97 1,EC11 1,1 EC10 1,1,SKP 1,1

580000, 50000 800000, 5000 800000, 5000 800000, 5000 800000, 5000 800000, 50000 800000, 50000

00624 002407		JMP	OBSAV		
00625'131120	EC50:	MOVEL	1,2		C-73
00626'151120		MOVZL	2,2		
00627'147000		ADD	2,1		
00630'125120		MOVEL	1,1		
00631'107000		ADD	0,1		
00632'001400		JMP	0,3		
00633'000000	DBSAV:	Ø	0,0		
00634'000000	EC10:	0			
00635'000000	ECI1:	Ø			
		"+			
00636'000053	EC20:				
00637 000055	EC21:				
	TO OUD		TT DYTE	IS A DICIT	
		235. 2012 - MARCH		IS A DIGIT	
	J& REDU		BINARY	17 11 15	
	,	JSR 0.CH			
	3	RETUR	entropy of the second of	IF NOT DIGIT	
	;	"		" IS "	
	; INPUT:				
	JOUTPUT				
	IDESTRO	YED: AC1			
	3				
00640'024412	CHEK:	LDA	1,MSK1		
00641 123400		AND	1,0		
00642'024412		LDA	1,19		
00643'122032		ADCZ#	1,0,SEC		
00644'001400		JMP	0.3		
00645'024406		LDA	1,NØ		
00646'106032		ADCZ#	Ø,1,52C		
00647 001400		JMP	0,3		
00650'122400		SUB	1,0		
00651 001401		JMP	1,3		
00652'000177	MSK1:	177			
00653'000060	NØ:	"0			
00654'000071	N9:	"9			
	1				
	ROUTINE	E TO GET	AN ALPHA	ANUMERIC STRING FROM	
	IKEYBOAN	RD AND ST	ORE IT	IN BYTE FORMAT WITH	
	JA TERMI	INATING 2	ERO BYTE	E	
	3				
	3	JSR e.WC	Conception of the second		
	3	ADDR (ADDRESS	TO PUT STRING)	
	3				
	JINPUT:	FIRST CH	ARACTER	IN ACØ	
	JALL ACC	CUMULATOR	RS ARE LO	DST	
	3				
00655'031400	WORD:	LDA	2,0,3	;ADDR TO PUT STRING	
00656'175400		INC	3,3		
00657 054446		STA	3.WOSAV		
00660'151120		MOVEL	2,2	BYTE POINTER	
00661 050445		STA	2,TWP		
00662 030445		LDA	2.MAXCS		
00663 050445		STA	2.TRAP		
00664'030442	MIKE:	LDA	2.TWP		
00665'010441		ISZ	TWP		
00666'024436		LDA	1.CR		
00667 106415		SUB#	0.1. SNR		
00670 000416		JMP	END1		
00671'155220		MOVER	2,3		
00672 031400		LDA	2,0,3	JOLD WORD	

00673'02443	6	LDA	1,MSKL		
00674'15100	2	MOV	2,2,SEC	INHICH BY	TE?
00675'15130	Ø	MOVS	2,2		
00676'13340	Ø	AND	1,2		
00677 11300	Ø	ADD	0,2	INEW BYTE	CODA KON (SEARS)
00700'15100	2	MOV	2,2,520		
00701'15130	0	MOVS	2,2	SWAP BAC	K
00702'05140	0	STA	2,0,3	; PUT BACH	C READOWN PERSON
00703'01442	5	DSZ	TRAP		
00704'00041	5	JMP	MARK		
00705'03042	1	LDA	2,TWP		
00706'15522	0 END1:	MOVER	2,3	JPUT Ø IN	LAST BYTE
00707 03140	0	LDA	2,0,3		
00710'15100	5	MOV	2,2,SEC		
00711'00040	4	JMP	LEFT		
00712'15240	Ø	SUB	2,2		
00713'05140	Ø	STA	2,0,3		
00714'00241	1	JMP	EWOSAV		
00715'02400	45 LEFT:	LDA	1. MSKR		
00716'13340	Ø	AND	1,2		
00717'05140	Ø	STA	2,0,3		
00720.00240	5	JMP	EWOSAV		
00721'00601	5- MARK:	JSR	e.GETT		
00722'00600	3-	JSR	0.PRN2		
00723'00074	1	JMP	MIKE		
00724'00001	5 CR:	15			
00725'00000	Ø WOSAV:	Ø			
00726'00000	Ø TWP:	0			
00727 00002	Ø MAXCS:	20			
00730'00000	Ø TRAP:	0			
00731 17740	Ø MSKL:	177400	JL.H. MA	SK	
		. END			

1	-					C-75
			.TITL	LOADS		
			.ENT	. HEAVY		
			•EXTD	.NUM	GETT, .DBIN, .MESS	
			•EXTD	*PRN2	PAGE	
			.EXTN	CONTR		
			.ZREL			
	00000-000000.	.HEAVY:	LOADS			
			.NREL			
		1				
		I ROUT	INE TO M	ULTIPLY (OR DIVIDE ALL BLOCK	
				AS) BY A		
		3				
	00000 054526	LOADS:	STA	3,RTRN	ISAVE ALL AC'S	
	00001 '040526		STA	Ø,ZER	an and a second a se	
	00002'044526		STA	1. ONE		
	00003 050526		STA	2,TWO		
	00004'0060075		JSR	8.PAGE		
	00005.0060055		JSR	e.MESS		
	00006'000155'		MS02	A		
	00007'177324		-300.			
	00010'001130		600.			
		1				
		1	CHECK F	OR MULT	DIV	
		1			and a start of the start and	
	00011 0060055	The part of	JSR	e.MESS		
	00012'000172'		MSØ4			
	00013'000113		75.			
	00014'000702		450.			
	00015'0060035	012:	JSR	e.GETT		
	00016'040514	0.011.0	STA	Ø,DIG	JSTORE M OR D	
	00017'024514		LDA	1.MM	STORE II ON D	
	00020'106415		SUB#		JIS IT M ?	
	00021'000411		JMP	OUT		
	00022'024512		LDA	1,DD	J IS IT D	
	00023'106415		SUB#	Ø,1,SNR		
	00024'000406		JMP	OUT		
	00025'0060055		JSR	0.MESS		
	00026'000227'		MSØ5	eencoo		
	00027'000310		200.			
	00030'000651		425.			
	00031'000764		JMP	OVR		
	00032'0060065	OUT:	JSR	0.PRN2		
	00033'152400	001.	SUB	2,2		
	00034'050504		STA	2,WHER		
	00035 024476		LDA	1.MM		
	00036'106415		SUB#	Ø,1,SNR		
	00037 000403		JMP	PAST		
	00040'152520		SUBEL	2,2		
	00041 050477		STA	2,WHER		
	00041 030411		STA	COWNER		
			GET CON	STANT		
			GET CUN	STANT		

1				
00042'0060055 PAST:	JSR	e.MESS		
00043'000237'	MSØ6			
00044 000226	150.			
00045'000567	375 -			
00046'0060045	JSR	e.DBIN		NAME AND ADDRESS TO LOT
00047 044472	STA	1.CNST	ISTORE	CONSTANT
1				
1	HERE WI	E GO !		

0	7	1
10.00	- 1	h
	- 1 .	U

	3				
00050'0340025		LDA	3, .M1	JGET 1ST B	LOCK POINTER
00051 054464		STA	3.BLK		
00052'0240015		LDA	1 NUM	JGET NO. O	F BLOCKS
00053'044463		STA	1.CNT		
00054'031400	OVR2:	LDA	2,0,3		
00055'050462	00000000	STA	2. TEMP	ISAVE FOR	LATER
00056'021014		LDA	0,14,2	JGET AREA	
00057'101005		MOV	the second se	ISKIP ERAS	ED BLOCK
00060'000425		JMP	TRAP	POILT LINE	20 02001
00061 024457		LDA	1. WHER		
00062'125004		MOV	and the second second	JIF NOT Ø	DIVIDE
		JMP	DIVD	JIF NOT 0	DIVIDE
00063'000412	MID To	-5-52.0	The Part of the		
00064'111000	MULT:	MOV	0,2		
00065'102400		SUB	0.0		
00066'024453		LDA	1, CNST		
00067 073301		MUL			
00070'030447		LDA	2, TEMP		
00071'045014		STA	and the strength of the second s	ISTORE NEW	
00072'125132		MOVEL#	C. Contract - C. C.	JTEST FOR	>11111
00073'000426		JMP FAIL	ALCONT ON ADDRESS		
00074'000411		JMP	TRAP		
00075'105000	DIVD:	MOV	0,1	JAREA IN A	
00076'102400		SUB	0.0	JCLEAR HI	PART
00077'030442		LDA	2.CNST		
00100'132432		SUBZ#	1,2,520	J DIV TEST	
00101'000420		JMP	FAIL		
00102'073101		DIV			
00103'030434		LDA	2, TEMP		
00104'045014		STA	1,14,2		
00105'010430	TRAP:	ISE	BLK		
00106'034427		LDA	3.BLK		
00107 014427		DSE	CNT		
00110'000744		JMP	OVR2	JDO NEXT B	LOCK
00111'020416		LDA	Ø,ZER		
00112'024416		LDA	1.ONE		
00113'030416		LDA	2.TWO		
00114'0060055		JSR	e.MESS		
00115'000252'		MSØ9			
00116'177160		-400.			
00117 000372		250.			
00120'002422		JMP	econ		
00121'0060055	FAIL:	JSR	e.MESS		
00122'000143'		MSØ8			
00123'177470		-200.			
00124'000310		200.			
00125 002415		JMP	econ		
00126'000000	RTRN:	Ø	COON		
00127 000000	ZER:	Ø			
00127-000000		0			
00131 000000	ONE:	Ø			
	TWO:	0			
00132'000000	DIG:				
00133'000115	MM:	"M			
00134'000104	DD:	"D			
00135'000000	BLK:	Ø			
00136'000000	CNT:	Ø			5444 mar #2 #3 #5
00137'000000	TEMP:	Ø			
00140'000000	WHER:	Ø			
00141 000000	CNST:				
00142'177777	CON:	CONTR			

	1		
00143'040506	MSØ8:	• TXT	*FA
00144'046111	IL		
00145'042105	ED		
00146'051454	,5		
00147 040524	TA		
00150'052122	RT		
00151'040440	A		
00152'020124	т		
00153'026520	P-		
00154'000061	1*		
00155'046102	MS02:	• TXT	*BL
	OC	•171	TOL
00156'041517			
00157'020113	K		
00160'042527	WE		
00161'043511	IG		
00162'052110	HT		
00163'046440	M		
00164'042117	OD		
00165'043111	IF		
00166'041511	IC		
00167'052101	AT		
00170'047511	IO		
00171 000116	N*		
00172 047504	M504:	•TXT	*D0
00173'054440	Y		
00174'052517	OU		
00175'053440	W		
00176 051511	IS		
00177'020110	н		
00200'047524	TO		
00201 046440	М		
00202'046125	UL		
00203'044524	TI		
00204'046120	PL		
00205'020131	Y		
00206'046450	CM		
00207 020051)		
00210'051117	OR		
00211 042040	D		
00212'053111	IV		
00213'042111	ID		
00214'020105	E		
00215'042050	(D		
00216'020051	>		
00217'044124	TH		
00220 020105	E		
00221'042527	WE		
00222'043511	IG		
00223'052110	нт		
00224'020123	S		
DRELA DEDILO	-		

00225'020077 ? 00559,000000 * 00227 052515 MS05: ST 00230 052123 00231 041040 B 00232 020105 E 00233 020115 M 00234'051117 OR 00235'042040 D

•TXT

*MU

				L-18
00236'000040	*			
00237'044127	M506:	• TXT	*WH	
00240'052101	AT			
00241'044440	I			
00242'020123	S			
00243'044124	TH			
00244'020105	E			
00245 040506	FA			
00246'052103	CT			
00247 051117	OR			
00250'037440	?			
00251 000040	*		Jan This.	
00252'047503	M509:	• TXT	*C0	
00253'050115	MP			
00254'042514	LE			
00255'042524	TE			
00256'026104	D,			
00257 053440	W			
00260'044501	AI			
00261 044524	TI			
00262'043516	NG			
00263'040040	e			
00264'041440	C			
00265'047117	ON			
00266'051124	TR *			
00267 '000000		.END		
		+ END		

		•TITL	FORD		
			MENT LAW	FOR ALL	
	JCONTAC	T POINTS			
		•EXTD		. NUM. EMPT.	
		• EXTD	.VEC.SI	CAL, PLTS, S	PRP. PRES
		• EXTD	.MESS	GETT IPRN	
		•EXTD	.ROT. UI	REP. TREC	
		.EXTD	.NVEC	PAGE . ALPH	HEAVY
		.EXTN	CONTR		
		.ENT	.FORD	TIME,MU	
		. ZREL			
00000-000000	MU:	000000	FRICTI	ON COEF. (DE	FAULT VALUE = .0)
00001-000033'	.FORD:	FORD			A CONTRACTOR OF THE OWNER OF THE
00002-000001	.KDN:	1	INORMAL	DAMPING FAC	TOR
00003-000001	.KDS:	1	SHEAR I	DAMPING FACT	OR
00004-000000	XCP:	0			
00005-000000	YCP:	Ø			
00006-000000	DELS:	Ø			
00007-000000	DELN:	Ø			
00010-000000	FN:	Ø			
00011-000000	FDSAV:	Ø			
00012-000000	LOCPR:	0			
00013-000000	LOCBL:	Ø			
00014-000000	LOCBP:	Ø			
00015-000000	OLINK:	Ø			
00016-000000	COUNT:	Ø			
00017-000000	PRLNK:	Ø			
00000-000000	cos:	Ø			
00021-000000	SIN:	Ø			
00022-000000	COSF:	Ø			
00023-000000	SINF:	Ø			
00024-000672	.TIME:	DYNFAC			
		.NREL			
00000'102440	MULS:	SUBO	0.0		
00001'050420		STA	2, SV2		
00002'027400		LDA	e1,0,3	JA	
00003'033401		LDA	82,1,3	18	
00004'125112		MOVL#	1,1,SZC		
00005'124460		NEGC	1,1		
00006'151112		MOVL#	2,2,SEC		
00007 150460		NEGC	2,2		
00010'073301		MUL			
00011 0300055	5	LDA	2. MSKR		
00012'143700		ANDS	2,0	JTAKE MIDDL	E 8 BITS
00013'125300		MOVS	1.1		
00014'147400		AND	2,1		
00015'107002		ADD	0,1,SEC		
00016'124400		NEG	1,1		
00017 030402		LDA	2, SV2		
00020'001402		JMP	2,3	JA*B IN AC1	
000021 .000000	SV2:	Ø			

XDL: 00022.000000 YDL: 00023'000000 00024'000000 XDP: 00025.000000 YDP: 00026.000000 DAP: 00027 000000 DAL: DXL: 00030 000000 00031 000000 DYL:

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	3			C-80
00032.000310.	NEXTR:	NEXTB		0 00
00033'054011-	FORD:	STA	3.FDSAV	
00034'0340025		LDA	3 M5	; INITIAL PROD POINTER
00035'054012-		STA	3,LOCPR	
00036'054015-		STA	3.OLINK	
00037 0200035		LDA	ØNUM	
00040'040016-		STA	Ø, COUNT	
00041 '0340015		LDA	3 M1	JINITIAL BLOCK DAT. PNTR.
00042'054013-		STA	3,LOCBL	
00043 036012-	LOOP:	LDA	3, PLOCPI	R JIST WORD
00044'175112	ENTRY:	MOVL#	3,3,SEC	JLIST TAIL FLAG?
00045 002765		JMP	ENEXTR	JYES, NEXT BLOCK
00046'054017-		STA	3.PRLNK	
00047 021400		LDA	ALL STREET STREET STREET	CONTROL WORD
00050 040023-		STA		JSIN FLAG IN BIT Ø
00051 101100		MOVL	0,0	I PERSONAL INCOME.
00052'040022-		STA	Ø,COSF	JCOS FLAG IN BIT Ø
00053'021410		LDA	0,10,3	JSIN
00054'040021-		STA	Ø,SIN	
00055'021411		LDA	0,11,3	JCOS
00056'040020-		STA	Ø,COS	1005
00057 021412		LDA	0,12,3	
			Ø, XCP	JX CONTACT POINT
00060 040004-		Contraction of the second	0,13,3	TA CONTACT FOINT
00061'021413		LDA	Ø,YCP	JY CONTACT POINT
00062 040005-		STA		
000/01000010	JIO GEI	CONTRIBL		
00063'032013-		LDA	2,eLOCBI	
00064'021001		LDA	0,1,2	XG, THIS BLOCK
00065'024004-		LDA	1,XCP	
00066'106400		SUB	0,1	
00067 044733		STA	1.XDL	
00070'021003		LDA	0,3,2	IYG, THIS BLOCK
00071'024005-		LDA	1,YCP	
00072'106400		SUB	0,1	
00073'044730		STA	1,YDL	
00074'021022		LDA	0,22,2	
00075'040732		STA	Ø,DAL	
00076'004702		JSR	MULS	
00077'000027'		DAL		
00100.000053.		YDL	and the second	TALLA STATES
00101.051050		LDA		JDELTA-X, THIS BLOCK
00102'122400		SUB	1,0	ISUBTRACT ROT. CONTRIB.
00103'040725		STA	Ø,DXL	
00104'004674		JSR	MULS	
00105'000027'		DAL		
00106'000022'		XDL		
00107'021021		LDA	0,21,2	;DELTA-Y
00110'123000		ADD	1,0	
00111'040720		STA	Ø,DYL	
	3			
00112'034017-		LDA	3, PRLNK	
00113'021401		LDA	0,1,3	(NP:NB)
00114'0240055		LDA	1. MSKR	

00115'107400 00116'0300015 00117'133000 00120 050014-00121 031000 00122'021001

AND LDA ADD STA LDA LDA 0.1 JBLOCK # OF POINT 2. . MI 1,2 2,LOCBP ;DATA POINTER (POINT) 2,0,2 IXG, OTHER BLOCK 0,1,2

				C-8	1
00123'024004-		LDA	1,XCP		1
00124'106400		SUB	0.1		
00125'044677		STA	1,XDP		
00126'021003		LDA	0,3,2	JYG, OTHER BLOCK	
00127 024005-		LDA	1,YCP		
00130'106400		SUB	0,1		
00131'044674		STA	1,YDP		
00132'021022		LDA	0,22,2		
00133'040673		STA	Ø,DAP	JDELTA-ALPHA	
00134'004644		JSR	MULS		
00135'000026'		DAP			
00136'000025'		YDP			
00137 021020		LDA	0,20,2	JDELTA-X, NB(P)	
00140'122400		SUB	1.0		
00141 024667		LDA	1,DXL		
00142'122400		SUB	1.0	JDXP-DXL	
00143'040570		STA	Ø,DELX		
00144'004634		JSR	MULS		
00145'000026'		DAP			
00146'000024'		XDP			
00147'021021		LDA	0,21,2	;DYP	
00150'123000		ADD	1,0		
00151'024660		LDA	1,DYL		
00152'122400		SUB	1,0	JDYP-DYL	
00153'040561		STA	Ø,DELY		
00154'004562		JSR	TRANS	JTRANSFORMATION ROUTINE	
00155'030017-		LDA	2, PRLNK		
00156'021005		LDA	0,5,2	JOLD N (NORM. DISP.)	
00157'163000		ADD	3,0		
00160'041005		STA	The second s	INEW N	
00161'165000		MOV	3,1		
00162'030553		LDA	2.KN	INORMAL STIFFNESS	
00163'102400		SUB	0.0		
00164'125112		MOVL#	1,1,SZC		
00165'124400		NEG	1,1		
00166'073301		MUL			
00167 175113		MOVL#	3,3,SNC		
00170'124400		NEG	and the second	JINVERT ORIG. SIGN	
00171'030017-		LDA		; FOR +VE FN	
00172'021006		LDA		JOLD NORMAL FORCE, FN	
00173'125112		MOVL#	1,1,SEC		
00174'000405		JMP	OK		
00175'107000		ADD	0,1		
00176'125112		MOVL#	1,1,SEC		
00177'006506		JSR	eLM1 STOR		
00200'000404	OK:	JMP ADD	And the set of the set of	JADD IN INCREMENT	
00202 125112	UN.	MOVL#		JEERO ADHESION ASSUMED	
00203 000520		JMP	DELET	ISET FORCES TO ZERO	
00204'045006	STOR:	STA	1,6,2	INEW NORMAL FORCE	
00205'044010-	Ston.	STA	1.FN		
00206'165000		MOV	3,1		
00000 100000				DEVENUE FARTOR	

00207'030002-00210'102400 00211'125112 00212'124400 00213'073301 00214'175113 00215'124400 00216'020010LDA SUB MOVL# NEG MUL MOVL# NEG LDA

2,.KDN JDAMPING FACTOR 0,0 1,1,S2C 1,1 3,3,SNC 1,1 0,FN

 -					
00217'123000		ADD	1,0	C-1	82
00220'125112		MOVL#	1,1,SEC		
00221 000403		JMP	NC		
00222'101112		MOVL#	0,0,SEC		
00223'006463		JSR	BLMØ		
00224'040510	NC:	STA	Ø, DELY		
	3	The state of the			
00225 030017-	-	LDA	2. PRLNK		
00226'006501		JSR	eSHR	GET SHEAR FORCE	
000000	1		C Stat	Design of the second	
00227 * 040504		STA	Ø,DELX		
00230'004506		JSR	TRANS		
00200 004000	JADD GLO	BAL FOR	CONTRACTOR OF THE OWNER	ING FROM	
	THIS CO			A ACCORDENCE MORNEY STREET	
00231 006453		JSR	EMOMT	MOMENT, THIS BLOCK	
00232'000007-		DELN	citorit	shonenty this becom	
00233'000006-		DELS			
00234'000022'		XDL			
00235'000023'		YDL			
00236'032013-		LDA	2, 0LOCBI	L STHIS BLOCK	
00237 021017		LDA	0,17,2	- FILLS DECON	
00240'122400		SUB	1.0		
00241 '041017		STA	0,17,2	INEW MSUM	
00242'021007		LDA	0,7,2	JOLD FXSUM	
00243'024006-		LDA	1.DELS	JOLD TROOM	
00244'123000		ADD	1,0		
00245'041007		STA	8,7,2	JNEW FXSUM	
00246'021016		LDA	0,16,2		
00247 024007-		LDA	1,DELN	JOLD 1 1001	
00250'122400		SUB	1.0		
00251'041016		STA		INEW FYSUM	
00252 006432		JSR	EMOMT	JULA FISON	
00253'000007-		DELN	enoni		
00254'000006-		DELS			
00255'000024'		XDP			
00256'000025'		YDP			
00257 '032014-		LDA	2, eLOCB	P JOTHER BLOCK	
00260'021017		LDA	0,17,2	A CANADA A C	
00261'123000		ADD	1,0	1000	
00262'041017		STA	1925 C	INEW MSUM	
00263'021007		LDA	0,7,2	JAS ABOVE, BUT	
00264'024006-		LDA		# WITH OPPOSITE SIGNS	
00265'122400		SUB	1,0		
00266'041007		STA	0,7,2		
00267 021016		LDA	0,16,2		
00270 024007-		LDA	1.DELN		
00271'123000		ADD	1,0		
00272'041016		STA	0,16,2		
00273'0200065		LDA		PLOT VECTORS IF FLAG	SET
00274'101004		MOV	0,0,SZR	E PROTA ANALANAMOTAN	
00275'006412		JSR	EVDISP		
00276'034017-	CHAIN:	LDA	3, PRLNK		
00277'171400		INC	3,2		
00300'151400		INC	2,2	JGET LINK ADDRESS	
00301'050015-		STA		FREVERSE LINK	
00302'035402		LDA	3,2,3	and the second sec	
00303'002425		JMP		JGET NEXT ENTRY	
00304'000432'	MOMT:	MOM			
00305'001143'	LM1:				
00306'001150'	LMØ:	LIMØ			
	and a second	Constanting of the second s			

							0.00
-							C-83
	00307 .000203.		VDIS				
	and the state of the second	JNEXT BI					
	00310'010012-	NEXTB:	ISZ	LOCPR	JINCR. PROD L	OCATOR	
	00311'034012-		LDA	3,LOCPR			
	00312'054015-		STA	3, OLINK			
	00313'010013-		ISZ	LOCBL	JINCR. DATA L	OCATOR	
	00314'014016-		DSZ	COUNT	JEXIT IF ALL I	BLOCKS	
	00315'002414		JMP	eloopr	J SCANNED		
	00316'0300125		LDA	2. PRES			
	00317'151112		MOVL#	2,2,SEC			
	00320'002011-		JMP	e FDSAV	INO PRESS. SEL	GMENTS	
	00321'002401		JMP	PRS	JGET FORCES FI	ROM PR.	SEGS.
	00322'000637'	PRS:	PRESU				
	00323'102400	DELET:	SUB	0,0			
	00324'041006		STA	0,6,2			
	00325'041007		STA	0,7,2			
	00326'000750		JMP	CHAIN			
	00327 '000553'	SHR:	SHEAR	CHAIN			
	00330'030044'		ENTRY				
		ENTR:					
	00331'000043'	LOOPR:	LOOP				
	00332'000000	SAVE:	Ø				
	00333'000000	DELX:	Ø				
	00334'000000	DELY:	Ø				
	00335'000003	KN:	3				
	00336'054774	TRANS:	STA	3.SAVE			
	00337'024774		LDA	1.DELX			
	00340'030020-		LDA	2,005			
	00341'102440		SUBO	0,0	JCLEAR CARRY		
	00342'125112		MOVL#	1,1,SEC			
	00343'124440		NEGO	1 . 1	SET CARRY		
	00344'073301		MUL		;DELX*COS		
	00345'125112		MOVL#	1,1,SEC	; ROUND UP IF N	VEC.	
	00346'101400		INC	0.0			
	00347'101002		MOV	0,0,SEC			
	00350'100400		NEG	0.0	IRESTORE SIGN		
	00351 024022-		LDA	1,COSF			
	00352'125102		MOVL	1,1,SEC			
	00353'100400		NEG	0,0			
	00354'115000		MOV	0,3	JPARTIAL SUM	IN AC3	
	00355'024757		LDA	1.DELY			
	00356'030021-		LDA	2,SIN			
	00357'102440		SUBO	0,0			
	00360'125112		MOVL#	1,1,SEC			
	00361'124440		NEGO	1,1			
				1.51	JDELY*SIN		
	00362'073301		MUL	1.1.570	JROUND UP IF N	VEC.	
	00363'125112		MOVL#		SHOUND OF IF I	.20.	
	00364'101400		INC	0,0			
	00365'101002		MOV	0,0,SEC			
	00366'100400		NEG	0,0			
	00367 024023-		LDA	1,SINF			
	00370'125102		MOVL	1,1,SEC			
	000711100400		ALCO	0.0			

00371'100400 00372'117000 00373'054006-00374'024740 00375'030020-00376'102440 00377'125112 00400'124440 00401'073301 NEG Ø, ADD Ø, STA 3, LDA 1, LDA 2, SUBO Ø, MOVL# 1, NEGO 1, MUL

0,0 0,3 3,DELS 1,DELY 2,COS 0,0 1,1,SZC 1,1

JDELX*COS+DELY*SIN

JDELY*COS

					, 0
00402'125112		MOVL#	1,1,SEC	JROUND UP IF NEC.	
00403'101400		INC	0.0		
00404'101002		MOV	0,0,SZC		
00405'100400		NEG	0,0		
00406'024022-		LDA	1,COSF		
00407'125102		MOVL	1,1,SZC		
00410'100400		NEG	0,0		
00411'115000		MOV	0,3	PARTIAL SUM IN AC3	
00412'024721		LDA	1.DELX	THATTAL BOIT IN A05	
00413'030021-		LDA	2,SIN		
00414'102440		SUBO	0,0		
00415'125112		MOVL#	1,1,SZC		
00416'124440		NEGO	1,1		
00417 073301		MUL		JDELX*SIN	
00420'125112		MOVL#	1.1.520	ROUND UP IF NEC.	
00421 101400		INC	0,0	TROUND OF IT NEC.	
00422'101002		MOV	0,0,SZC		
00423'100400		NEG	0,0		
00423 100400		01 22/12	1,SINF		
INTERNAL // CONTRACTOR		LDA			
00425'125102		MOVL	1,1,SZC		
00426'100400		NEG	0,0	AND ALCON DOLLARD	
00427'116400		SUB	0,3	JDELY*COS-DELX*SIN	
00430'054007-		STA	3.DELN		
00431 002701		JMP	esave	AND TOUNDATES	
	States and March States			. AND TRUNCATES	
		A second s	15 01 32	2 BIT NUMBER	
		PUT: ACI			
00432'054444	MOM:	STA	3,TEMP	1010 - 10 - 194-194 194 1	
00433'027400		LDA		JA	
00434'033402		LDA	02,2,3	\$ XD IF	
00435'176400		SUB	3,3		
00436'125112		MOVL#	1,1,SZC		
00437 157000		ADD	2,3		
00440'151112		MOVL#	2,2,SEC		
00441 137000		ADD	1,3		
00442'102400		SUB	0.0		
00443'073301		MUL	1		
00444'162400		SUB	3,0	TRA	
00445'040432		STA	Ø,HI	JA*XDIF IN AC0:AC1	
00446'044432		STA	1,10		
00447 034427		LDA	3,TEMP	STREET STREET STREET STREET	
00450'027401		LDA	01,1,3	3B	
00451 033403		LDA	82,3,3	JYDIF	
00452'176400		SUB	3,3		
00453'125112		MOVL#	1,1,SZC		
00454'157000		ADD	2,3		
00455'151112		MOVL#	2,2,520		
00456'137000		ADD	1,3		
00457'102400		SUB	0.0		
00460'073301		MUL			
00461 162400		SUB	3.0	B*YDIF IN AC0:AC1	
00462'030415		LDA	5'HI		
00463'034415		LDA .	3,10		
00464'167022		ADDZ	3,1,SEC	JADD 2 D.P. NUMBERS	
00465'151400		INC	2,2		
00466'143000		ADD	2,0	JD.P. ANSWER IN ACO:A	C1
00467 '0300055		LDA	2. MSKR	INOW TAKE ONLY MIDDLE	
00470'143700		ANDS	2,0	I S BITS	
00471'125300		MOVS	1,1		
00472'147400		AND	2,1		

	C-85
0,1 3,TEMP	JRESULT IN AC1
4,3	RETURN TO CALL +5
3,VEC3	VECTOR PLOTTING ROUTINE
	X CONTACT POINT
1.YCP	3Y "
e.PLTS	SIST END (BEAM OFF)
1,DELS	
1 . XNUM	
e.SCAL	ISCALE FORCE FOR PLOTTING
Ø, XCP	
1,0	
Ø, XVEC	;X VECTOR
1.DELN	
1.YNUM	
e.SCAL	
Ø,YCP	
1.0	
0.1	IY VECTOR

00474 034402 LDA 00475'001404 JMP 00476'000000 TEMP: Ø 00477 0000000 0 HI: 00500 000000 LO: ø 00501 000000 XNUM: Ø 00502'000000 YNUM: Ø 00503'054446 VDIS: STA 00504 020004-LDA 00505 024005-LDA 00506'0060105 JSR 00507 000000 Ø 00510 024006-LDA STA 00511 044770 00512'006007\$ JSR 00513'020004-LDA 00514'123000 ADD 00515 040435 STA 00516'024007-LDA 00517 044763 STA JSR 00520'0060075 00521 020005-LDA SUB 00522'122400 00523'105000 MOV 0.1 JT VECTOR Ø, XVEC 00524'020426 LDA ; PLOT VECTOR 00525'0060105 JSR e.PLTS 00526'000001 ; BEAM ON 1 e.ALPH JSR 00527 0060235 2, NVEC ; TO PRINT VALUES 00530'0300215 LDA 2,2, SNR ;0=DONT PRINT 00531 151005 MOV JMP eVEC3 00532'002417 LDA Ø, XNUM 00533'020746 ; PRINT X JSR e.IPRN 00534 0060155 00535'000005 5 LDA Ø, YNUM 00536'020744 JSR JPRINT Y e.IPRN 00537 0060155 5 00540'000005 2, NVEC JIF>1, HALT FOR CHECK 00541 0300215 LDA MOVER 2,2,SER 00542'151224 JWAIT FOR ANY KEY JSR WAIT 00543'004402 00544'002405 eVEC3 JMP SKPDN 00545'063610 WAIT: TTI 00546 000777 JMP . - 1 00547 060210 NIOC TTI JMP 0,3 00550'001400 VEC3: 00551 000000 Ø XVEC: 00552 000000 Ø 3 JTHE FOLLOWING ROUTINE COMPUTES SHEAR FORCE

ADD

FROM SHEAR DISP. AND NORMAL FORCE. JIT ALSO ADDS IN DAMPING TERM, IF CONTACT IS INOT SLIDING.

SHEAR: 2,SVS2 00553'050455 STA 1,0,2 00554 025000 LDA Ø, FRMSK JTYPE # MASK 00555'020455 LDA 0,1,SER ; IF ZERO, USE DEFAULT 00556'107704 ANDS GETFR 00557 000454 JMP JFRICTION COEF (<1) 2,MU 00560 030000-LDA

00473'107000

00561 '024010-	SLIP:	LDA	1.FN	
00562'102400		SUB	0.0	
00563'073301		MUL		FN*MU IN ACO
00564 040443		STA	Ø.FSMAX	MAX POSS SHEAR FORCE
00565'030444		LDA	2.KS	; SHEAR STIFFNESS
00566 024006-		LDA	1,DELS	JINCR. SHEAR DISP.
00567'102440		SUBO	0.0	JCLEAR CARRY
00570'125112		MOVL#	1,1,SEC	
00571'124440		NEGO	1,1	SET CARRY IF DELS -VE
00572 073301		MUL		;DELS*KS (=DELTALFS])
00573'125002		MOV	1,1,SEC	
00574'124400		NEG	1,1	;RETURN SIGN
00575'030433		LDA	2, SVS2	
00576'021007		LDA	0,7,2	FS(OLD)
00577'107000		ADD	0.1	JRAW FS
00600 044426		STA	1,FS	
	3			
	; THE	FOLLOWING	LINE WAS	S IN ERROR IN PAC'S
00601 045007		STA	1,7,2	:7/30/76 ERROR FOUND
	3			LOS ATTRACT ALTER
00602'121102		MOVL	1,0,SEC	
00603'124400		NEG	1.1	
00604'020423		LDA	Ø,FSMAX	
00605'122513		SUBL#	1.0.SNC	JEXCEEDED MAX?
00606'000405		JMP	DAMP	SNO. ADD IN DAMPING
00607'125002		MOV	1,1,SZC	SIGN?
00610'100400		NEG	0.0	
00611'041007		STA	0,7,2	JNEW FS IN ACO
00612.001400		JMP	0.3	JEXIT
00613 024006-	DAMP:	LDA	1,DELS	
00614'030003-		LDA	2 KDS	JDAMPING FACTOR
00615'102440		SUBO	0.0	
00616'125112		MOVL#	1,1,SEC	
00617'124440		NEGO	1,1	
00620'073301		MUL		
00621'125002		MOV	1,1,SEC	
00622'124400		NEG	1 . 1	
00623'020403		LDA	Ø,FS	
00624'123000		ADD	1.0	JADD IN DAMPING FORCE
00625'001400		JMP	0.3	JEXIT (OUTPUT: ACO)
00626'000000	FS:	Ø		
00627 '000000	FSMAX:	Ø		
00630'000000	SVS2:	Ø		
00631 '000003	KS:	3	ISHEAR S	STIFFNESS
00632'017400	FRMSK:	17400	IMASK FO	OR TYPE # PART OF CONT. WORD
00633'0300115	GETFR:	LDA	2. SPRP	
00634'133000		ADD	1,2	
00635'031000		LDA	2,0,2	JGET APPROPRIATE FRICTION
00636'000723		JMP	SLIP	

JTO ADD IN PRESSURE FORCES FROM LINKED

3,.M1 0,3 3,0,3

1,.MSKR 1,0 JNB

0,0,2

JBLOCK POINTER

JLIST OF PRESSURE SEGMENTS.

00637 '021000	PRESU:	LDA	
00640'0240055		LDA	
00641'123400		AND	
00642'0340015		LDA	
00643'117000		ADD	
00644'035400		LDA	

1----

3

						C.C.T.R. CO.
00645'021003		LDA	0,3,2	JM I	NCREMENT	
00646'025417		LDA	1,17,3		MSUM	
00647 107000		ADD	0,1		and and a second se	
00650'045417		STA	1,17,3	INEW	MSUM	
	3		1000			
00651 021004		LDA	0,4,2	JFX	INCREMENT	
00652'025407		LDA	1,7,3	JOLD	FXSUM	
00653'107000		ADD	0,1			
00654'045407		STA	1,7,3	INEW	FXSUM	
	;					
00655'021005		LDA	0,5,2	FY	INCREMENT	
00656'025416		LDA	1,16,3	;OLD	FYSUM	
00657 107000		ADD	0,1			
00660'045416		STA	1,16,3	INEW	FYSUM	
	;					
00661 031002		LDA	2,2,2	;LIN	К	
00662'151115		MOVL#	2,2, SNR			
00663 000754		JMP	PRESU			
00664'002011-		JMP	eFDSAV	JEND	OF CHAIN.	12112
	3		-			
	J ROUTI	NE TO CH	ANGE TRE	C, ET	C •	
	1					
00665'000040	DTREC:	40				
00666'000001	DKDN:	1				
00667 000012	DKDS:	12				
00670'000140	DROT:	140				
00671 000023	DUREP:	23				
	3					
00672'0060225	DYNFAC:	JSR	e.PAGE			
00673'0060235		JSR	e.ALPH			
00674'0060135		JSR	e.MESS			
00675'001212'		DMSØ				
00676 177470		-500.				
00677 001320		720.				
00700'0060135		JSR	e.MESS			
00701'001234'		DMS1				
00702'177665		-75.				
00703'001236		670.				
00704 0060135		JSR	e.MESS			
00705'001244'		DMS2				
00706'000175		125.				
00707 001200		640.				
00710'0200205		LDA	Ø. TREC	JTIM	E STEP	
00711'0060155		JSR	e.IPRN			
00712'000004		4				
00713'0060135		JSR	e.MESS			
00714'001250'		DMS3				
00715 000175		125.				
00716'001130		600.				
00717'020002-		LDA	Ø. KDN	INOR	MAL DAMPIN	G FAC
0070010060155		150	A TPPN			

00720'0060155 00721'000004 00722'0060135 00723'001254' 00724'000175 00725'001060 00726'020003-00727'0060155 00730'000004 00731'0060135 JSR 4 JSR DMS4 125. 560. LDA JSR 4 JSR

Ø,.KDS e.IPRN e.MESS

e.IPRN

e.MESS

ISHEAR DAMPING FAC

				C-
00732'001260'		DMS5		
00733'000175		125.		
00734'001010		520.		POT THE FAC
00735'0200165		LDA	Ø, .ROT	;ROT. TIME FAC
00736'0060155		JSR	e.IPRN	
00737 000005		5		
00740 0069135		JSR	e.MESS	
00741 001264		DMS6		
00742'000175		125.		
00743'000740		480 .		
00744'0200175		LDA	Ø. UREP	JUPDATE COUNTER
00745'0060155		JSR	e.IPRN	
00746'000004		4		
	3			
00747'0060135		JSR	e.MESS	
00750'001270'		DMS7		
00751 177470		-200.		
00752'000536		350.		
00753'0060135		JSR	0.MESS	The south a state of the second
00754'001306'		DMS8		
00755'000454		300.		
00756'000454		300.		
00757 0060135		JSR	e.MESS	
00760'001325'		DMS9		
00761 000454		300.		
00762 000404		260.		
00763'0060135		JSR	e.MESS	
00764'001367'		DM10		
00765 000454		300.		
00766'000334		220.		
00767 0060135		JSR	e.MESS	
00770'001344'		DMS10		
00771 '000454		300.		
00772'000264		180.		
00112 000001	1			
	I GET	CONTROL H	KEY	
	1			
00773'0060145		JSR	@.GETT	
00774'024414		LDA	1.WCHR	JIS IT A W
00775'106415		SUB#	0.1. SNR	
00776 0060245		JSR	e.HEAVY	JYES
00777 024407		LDA	1, ICHR	JIS IT AN I?
01000'106415		SUB#	Ø.1.SNR	
01001'000410		JMP	UP	JYES
01002'024405		LDA	1.DCHR	JIS IT A D ?
01003'106415		SUB#	Ø.1.SNR	
01004'000434		JMP	DWN	JYES
01005'002535		JMP	econ	JNONE-GO TO CONTR
01006'000111	ICHR:	"1	and the second se	performance + provide
01007 000104	DCHR:	"D		
01010'000127	WCHR:	••W		
01011 020002-	100000000000000000000000000000000000000	LDA	Ø. KDN	

01011'020002- UP: 01012'024654 01013'106432 01014'000521 01015'122400 01016'040002-01017'0200205 01020'024645 01021'122400

-

LDA LDA SUBZ# JMP SUB STA LDA LDA SUB 1,DKDN 0,1,SEC ;IFKDN=DKDN ALREADY AT MAX MAX 1,0 0,.KDN 0,.TREC 1,DTREC 1,0

1	01022'0400205		STA	Ø. TREC
	01023'020003-		LDA	Ø. KDS
	01024'024643		LDA	1.DKDS
	01025 122400		SUB	
	01026'040003-			1,0
			STA	Ø. KDS
	01027'0200165		LDA	Ø ROT
	01030'024640		LDA	1, DROT
	01031'122400		SUB	1,0
	01032'0400165		STA	Ø. ROT
	01033'0200175		LDA	Ø. UREP
	01034'024635		LDA	1.DUREP
	01035'122400		SUB	1,0
	01036'0400175		STA	Ø,.UREP
	01037 000426		JMP	OUTPT
		3		
	01040'0200205	DWN:	LDA	Ø. TREC
	01041'024624		LDA	1.DTREC
	01042'107000		ADD	0,1
	01043'0440205		STA	1. TREC
	01044'020002-		LDA	Ø. KDN
	01045'024621		LDA	1.DKDN
	01046'107000		ADD	Ø.1
	01047 044002-		STA	1KDN
	01050 020003-		LDA	Ø. KDS
	01051'024616		LDA	1, DKDS
	01052'107000		ADD	0,1
	01053'044003-		STA	1KD5
	01054'020016\$		LDA	Ø ROT
	01055'024613		LDA	1,DROT
	01056'107000		ADD	0,1
	01057'0440165		STA	1,.ROT
	01060'0200175		LDA	Ø UREP
	01061'024610		LDA	1, DUREP
	01062'107000		ADD	0.1
	01063'0440175		STA	1. UREP
	01064'000401		JMP	OUTPT
		3		
	01065'0060135	OUTPT:	JSR	e.MESS
	01066'001361'		DMS11	
	01067 176701		-575.	
	01070.001236		670.	
	01071 0060135		JSR	e.MESS
	01072'001244'		DMS2	
	01073'001161		625.	
	01074'001200		640.	
	01075'0200205		LDA	Ø. TREC
	01076'0060155		JSR	e.IPRN
	01077 000004		4	
	01100'0060135		JSR	e.MESS
	01101'001250'		DMS3	
	01101 001100		625.	

01102'001161 01103'001130 01104'020002-01105'0060155 01106'000004 01107'0060135 01110'001254' 01111'001161 01112'001060 01113'020003625. 600. LDA 0,.KDN JSR 0.IPRN 4 JSR 0.MESS DMS4 625. 560. LDA 0,.KDS

				C-90
01114'0060155		JSR	e.IPRN	C-30
01115'000004		4		
01116'0060135		JSR	0.MESS	
01117'001260'		DMS5		
01120'001161		625.		
01121 001010		520.		
01122.0200165		LDA	Ø ROT	
01123'0060155		JSR	e.IPRN	
01124'000005		5		
01125'0060135		JSR	e.MESS	
01126'001264'		DMS6		
01127'001161		625.		
01130'000740		480.		
01131'0200175		LDA	Ø. UREP	
01132'0060155		JSR	e.IPRN	
01133'000004		4		
01134'002406		JMP	econ	
	3			
	3			The second second second
01135'0060135	MAX:	JSR	e.MESS	
01136'001172'		ERR		
01137 177470		-200.		
01140'000226		150.		
01141'002401		JMP	econ	3 GO BACK TO CONTR
01142'177777	CON:	CONTR		
	;			
01143'054411	LIM1:	STA	3, RETN	
01144'004412		JSR	WARN	
01145'024410		LDA	1,LIMIT	
01146'034007-		LDA	3.DELN	
01147'002405		JMP	eretn	
01150'054404	LIMØ:	STA	3, RETN	
01151'004405		JSR	WARN	
01152'020403		LDA	Ø,LIMIT	
01153'002401		JMP	eretn	
	3	A. 1 (C.)		
01154'000000	RETN:	Ø		I BADACARI
01155'077777	LIMIT:	77777	IMAX NOR	MAL FORCE
01156'054413	WARN:	STA	3,RETR	
01157'006013\$		JSR	e.MESS	
01160'001404'		MW1		
01161'001522		850.		
01162'001332		730.		
01163'0060135		JSR	0.MESS	
01164'001412'		MW2		
01165'001522		850.		
01166'001313		715.		
01167'034402		LDA	3,RETR	
01170'001400		JMP	0,3	
01171'000000	RETR:	Ø		

01172'047523 01173'051122 RR 01174'026131 Y, 01175'046101 AL 01176'042522 RE 01177'042101 AD 01200'020131 Y 01201'052101 AT

3

01202'046440	M				C-91
01203'054101	AX				0-91
01204'046511	IM				
01205'046525	UM				
01206.053040	V				
01207 '046101	AL				
01210'042525	UE				
01211'000123	S*				
01212'027056	DMSØ:	.TXT	*		
01213'027056	**				
01214'027056	S				
01215'020056					
01216'054504	DY				
01217 040516	NA				
01220'044515					
01220 044313	MI				
	C				
01222'040520	PA				
01223'040522	RA				
01224'042515	ALCONTRACT,				
01225'042524	TE				
01226'051522					
01227 027056	••				
01230'027056	• •				
01231 027056	••				
01232'027056					
01233'000000			Victoria Maria		
01234'051120		•TXT	*PR		
01235'051505	ES				
01236'047105					
01237'020124					
01240'040526	VA				
01241'052514	LU				
01242'051505	ES				
01243'000000	*				
01244'052056	DMS2:	• TXT	*•T		
01245'042522	RE				
01246'020103	C				
01247'000075	= *				
01250'045456	DMS3:	•TXT	*•K		
01251 047104	DN				
01252'036440	=				
01253 000000	*				
01254'045456	DMS4:	• TXT	*•K		
01255'051504	DS				
01256'036440	=				
01257 000000	*				
01260'051056	DMS5:	•TXT	*•R		
01261 052117	OT				
01262'036440	=				
01263'000000	*				
01264'052456	DMS6:	• TXT	* • U		
01265'042522		- substanting	1. 1. 1. 1. 1. 1. 1. 1. 1. 1. 1. 1. 1. 1		
01266.020120	P				

.TXT

*FO

	-	-	-	

01267'000075 =* 01270'047506 DMS7: 01271'051125 UR 01272'047440 0 01273'052120 PT 01274'047511 I0 01275'051516 NS

01276'040440	A			THE REAL PROPERTY.	2011
01277'040526	VA			C-93	2
01300'046111	IL				
01301'041101	AB				
01302'042514	LE				
01303'026440	-				
01304'026455					
01305'000040	*				
01306'054524	DMS8:	• TXT	*TY		
01307'042520	PE				
01310'044440	I				
01311 052040	Т				
01312'020117	0				
01313'047111	IN				
01314'051103	CR				
01315'040505	EA				
01316'042523	SE				
01317'052040	T				
01320'046511	IM				
01321'020105	E				
01322'052123	ST			A DE LA PRESE DE LEO.	
	EP				
01323'050105	*				
01324'000000		TYT	*TY		
01325'054524	DMS9:	.TXT	*11		
01326'042520	PE				
01327'042040	D				
01330'052040	T				
01331'020117	0				
01332'042504	DE				
01333'051103	CR				
01334'040505	EA				
01335'042523	SE				
01336'052040	T				
01337'046511	IM				
01340'020105	E				
01341'052123	ST				
01342'050105	EP				
01343'000000	*	-			
01344'047101	DMS10:	•TXT	*AN		
01345'020131	Y				
01346'052117	OT				
01347'042510	HE				
01350'020122	R				
01351'042513	KE				
01352'020131	Y				
01353'020055	-				
01354'047516	NO				
01355'041440	С				
01356'040510	HA				
01357 043516	NG				
01360'000105	E*				
01361'042516	DMS11:	• TXT	*NE		
01362'020127	W				

DIGOL DEDIE! 01363'040526 VA 01364'052514 LU 01365'051505 ES 01366'000000 * 01367 054524 DM10: 01370.042520 PE 01371'053440

W

*TY .TXT

-

Т					C-93
0					0 50
MO					
DI					
FY					
W					
EI					
GH					
TS					
*					
MW1:	•TXT	*			
TO					
0					
HE					
AV					
Y*					
MW2:	• TXT	" *			
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**					
	0 M0 DI FY W EI GH TS * MW1: TO 0 HE AV Y* MW2: ** ** **	0 M0 DI FY W EI GH TS * MW1: •TXT TO 0 HE AV Y* MW2: •TXT ** ** **	0 M0 DI FY W EI GH TS * MW1: •TXT * TO 0 HE AV Y* MW2: •TXT * * ** **	0 M0 DI FY W EI GH TS * MW1: •TXT * TO 0 HE AV Y* MW2: •TXT " * ** ** ** ** ** ** ** ** **	0 M0 DI FY W EI GH TS * MW1: •TXT * TO 0 HE AV Y* MW2: •TXT * * ** **

. END

01420'000000

**

			C 04
		T 1 T 1	UPDAT C-94
		.TITL .ENT	.ALLB, .SING, .CPNT
		•EXTD	.M1,.M2,.M3,.M4,.M5,.M6,.M7,.MSKR
		.EXTD	.PON1, .PON2, .PRN1, .EMPT, .PSIE, .LENG
		.EXID	•TYP
		.EXTD	•MEM
		.ZREL	313 L 10 2 4 5 1 5 1 5 1 5 1 5 1 5 1 5 1 5 1 5 1 5
00000-00000	0. ALLB:	ALLB	
00001-00005	all and the second s	SING	
00002-00050	and the second second	CHA	POINTER TO WORD THAT CAN BE MODIFIED
00003-00000		Ø	
00004-00000	The second second	Ø	
00005-00000		Ø	
00006-00000		Ø	
00007-00000		0	
00010-00000		Ø	
00011-00000		Ø	
00012-00000		Ø	
00013-00000	New York data and the second	Ø	
00014-00000	10 L:	Ø	
		•NREL	
	ROUTIN		DATE ALL BLOCK CONTACTS
	;	JSR e.A	ALLB
	3		
00000 05441		STA	3,ALL3
00001 03400		LDA	3, •M1
00002'10240		SUB	0,0
00003 04041		STA SCAN	Ø,N88
	CONTRACTOR CONTRACTOR	STA	3,HOLD
00004'05441	and the second sec	LDA	2,0,3
00006'1510		MOV	2,2, SNR
00007 00240		JMP	BALLS INO MORE BLOCKS. EXIT!
00010'0244		LDA	1,NBB
00011'0044		JSR	SING JUPDATE SINGLE BLOCK CONTACTS
00012'01040		ISE	NBB
00013'03440		LDA	3,HOLD
00014'1754		INC	3,3
00015'0007	57	JMP	BEGIN
00016'0000	DO ALL3:	Ø	
00017'0000	MA NBB:	Ø	
00020.0000	00 HOLD:	Ø	
	3		THE THE PERMIT COMMENTER THE
	JAFTER	ALL SID	ES HAVE BEEN SCANNED, THIS
	ROUTIN	VE THROW	S OUT ALL ENTRIES IN CONTACT
- I Caleria			E NOT BEEN FLAGGED. 1,LBIT J"PRESERVE" FLAG
00021 0245		LDA	
00022'0340		LDA	3,.M5 Ø,NB
00023'0200		ADD	0,3 JLOCATOR OF CONTACT LIST
00024'1170		STA	3, OLINK JBACKWARDS LINK
00025'0544		LDA	3,0,3 JGET POINTER (OR -1)
00026'0354	DUONE.	MOVI #	3.3.5FC (FND?

00026'035400 00027'175112 PHONE: 00030'002500 00031'021400 00032'123415 00033'000410 00034'122400 00035'041400 00036'171400

MOVL# JMP LDA AND# JMP SUB STA INC 3,3,SEC JEND? @SIN3 JDONE. EXIT! Ø,0,3 J1ST WORD 1,0,SNR JIS PRESERVE FLAG SET DELET JNO, DELETE ENTRY 1,0 JKEEP ENTRYJ REMOVE FLAG 0,0,3 JPUT IT BACK 3,2

00037'151400	INC	2,2	JGET ACTUAL LINK ADDRESS
00040'050412	STA	2, OLINK	JREMEMBER REVERSE LINK
00041 035402	LDA	3,2,3	JGET NEXT ENTRY
00042'000765	JMP	PHONE	
	JTO DELETE	AN ENTRY, AND	PUT IT IN THE
	J"EMPTY" LI	ST.	
00043 0200145	DELET: LDA		GET LINK FROM LOCATOR
00044'054014S	STA	3. EMPT	PUT IN NEW LINK
00045'031402	LDA		
00046'041402	STA		STORE EMPT LINK IN IT
00047 052403	STA		
00050'155000	MOV		INEXT ENTRY
00051 000756	JMP	PHONE	J ENTRY
00052'000000	OLINK: Ø		
	J DOULT THE TO	UDDATE STAR	E DLOCK CONTACTS
			LE BLOCK CONTACTS
	J JSK	0.SING	
	INPUT: ACI	- BLOCK #	
		Construction of the control of the second se	START OF DATA, BLOCK NB
	1	101111211 10	
00053'054455	SING: STA	3,SIN3	
00054'044011-	STA	1,NB	
00055'021014	LDA	0,14,2	
00056'101005	MOV	0,0, SNR	
00057 002451	JMP	esin3	JZERO AREA. EXIT!
00060.051000	LDA	0,0,2	JCONTROL WORD
00061 0240105	LDA	1. MSKR	
00062'107400	AND		INO. OF POINTS
00063'044446	STA	a state of the second s	INEGATIVE POINT COUNTER
00064'126400	SUB		
00065'044012-	STA		ANT I PHOTU I THIS CIDE
00066'0060165	JSR		GET LENGTH L THIS SIDE
00067 040014-	STA		
00070'0060115	JSR		JGET GLOBAL CO-ORDS
00071'040441	STA		
00072'044441 00073'040003-	STA STA	Ø,XA	
00074'044004-	STA		
00075'024012-	LDA		
00076'000420	JMP		
00077'125400	BACK: INC		
00100'0060115	JSR		
00101 '040573	STA	Ø,XB	
00102'044573	STA	1.YB	
00103'050423	STA	2,AC2	
00104'004433	JSR	RED	SEARCH FOR CONTACTS
00105'030421	LDA		
00106'010012-	ISZ		
00107'024012-	LDA	A STATE OF MANAGEMENT	
00110'0060165	JSR		
00111 040014-	STA		INFW BECOMES OLD
001121020562	IDA	0.X8	ENEW BELUMED ULD

-8

00112'020562 00113'040003-00114'020561 00115'040004-00116'014413 DOWN: 00117'000760 00120'020412 00121'040553

LDA STA LDA STA DSZ JMP LDA STA 0,XB INEW BECOMES OLD 0,XA 0,YB 0,YA NPNTS IJUMP OUT IF DONE BACK 0,X0 ILAST LINE 0,XB

00122'020411		LDA	0,40	
00123'040552		STA	Ø,YB	
00124'004413		JSR	RED	JSEARCH FOR CONTACTS
00125'000674		JMP	SCAN	ISCAN FOR FLAGS
00126'000000	AC2:	Ø		
00127 '020000	LBIT:	20000		
00130'000000	SIN3:	0		
		70		
00131 000000	NPNTS:	Ø		
00132'000000	X0:	Ø		
00133'000000	YØ:	0		
00134'000000	XLBOX:	0		
00135'000000	YLBOX:	Ø		
00136'000000	XUBOX:	0	Law and	
				(XRANG, YRANG)
			YA), (XB,	(B)]
00137'054543	RED:	STA	3.SVR3	
00140'102520		SUBEL	0.0	
00141'040552		STA	0, BYPAS	JINITIALIZE SKIP FLAG
00142'030547		LDA	2,0100	
00143'020004-		LDA	Ø,YA	
00144'024531		LDA	1,YB	
00145'122512		SUBL#	1,0,SEC	JIS YA>=YB?
00146 000404		JMP	REV	INO
00147'044530		STA	1.YL	STORE YB AS LOWER
00150'040531		STA	Ø,YU	JYA AS UPPER
00151 000403		JMP	ON	
00152'040525	REV:	STA	Ø,YL	THE REVERSE
00153'044526		STA	1,YU	
00154'020003-	ON:	LDA	Ø,XA	
00155'024517		LDA	1,XB	
00156'122512		SUBL#		DO SAME FOR X
00157 000404		JMP	VER	For onite Fort it
00160'044516		STA	1, XL	
00161'040517		STA	Ø, XU	
00162'000403		JMP	ONN	
00163'040513	VER:	STA	Ø.XL	
00164'044514	VEN.	STA	1.XU	
00104 044314	FIND B	OX ADDRE:		
001/51004511	ONN:	LDA		
00165'024511	UNIN .	SUB	1,XL	
00166'102400			0.0	
00167'073101		DIV	0 0 570	
00170'101004		MOV	0,0,SZR	
00171'000405		JMP	+5	
00172'125005		MOV	1,1, SNR	
00173'000403		JMP	•+3	
00174'102520		SUBZL	0,0	
00175'106400		SUB	0,1	INO Y DOYES FROM OBLC
00176'044736		STA		INO. X BOXES FROM ORIG
00177'024500		LDA	1,YL	
00200'102400		SUB	0.0	
00201 073101		DIV		

0,0,SER

MOV

00203'000405 00204'125005 00205'000403 00206'102520 00207'106400 00210'044725 00211'024467 00212'102400

00202 101004

.+5 JMP MOV 1,1, SNR .+3 JMP 0.0 SUBZL SUB 0,1 STA 1, YLBOX INO. Y BOXES FROM LDA 1,XU 0,0 SUB

# m.				
00213'073101		DIV		
00214'044722		STA	1, XUBOX	INO. X BOXES FROM
00215'024464		LDA	1.YU	JORIGIN TO END
00216'102400		SUB	0,0	
00217'073101		DIV		
00220'020715		LDA	Ø,YLBOX	JNO. Y BOXES
00221'106400		SUB	0.1	INO. Y BOXES IN SCAN
00222'124000		COM	1 > 1	
00223'044463		STA	1,YRANG	JADD 1, MAKE -VE
00224'0340035		LDA	3. • M3	
00225'103120		ADDEL	0.0	MULTIPLY YLBOX BY 20
00226'103120		ADDZL	0.0	
00227'117000		ADD	0.3	
00230'024706		LDA	1,XUBOX	
00231 '020703		LDA	Ø, XLBOX	
00232'106400		SUB	0.1	JNO.X BOXES IN SCAN
00233'124000		COM	1 + 1	
00234'044451		STA	1, XRANG	
00235'044452		STA	1.XCNT	COPY FOR SCAN ROUTINE
00236'117000		ADD	0,3	START BOX ADDR IN AC3
00237 054445	LOOPO:	STA	3.NLEFT	JLEFT-HAND POINTER
00240'054443	LOOP:	STA	3.KEEP	JMOVING X POINTER
00241 035400		LDA	3,0,3	
00242'175112		MOVL#	3,3,SEC	;END MARK?
00243 000415		JMP	ENDM	JYES
00244'021400	THERE:	LDA	0,0,3	GET WORD IN LINKED LIST
00245'0300105		LDA	2. MSKR	
00246'11:3400		AND	0,2	JUST NB IN AC2
00247 024011-		LDA	1.NB	
00250'132415		SUB#	1,2, SNR	
00251 000404		JMP	MOVE	SAME BLOCK! DISCARD!
00252'054440		STA	3,5V3	
	3			
00253 004443		JSR	PUSH	; (NP:NB) IN ACO; HOME NB IN AC1
	;			
00254'034436		LDA	3, SV3	1
00255'035401	MOVE:	LDA	3,1,3	J2ND WORD (=LINK)
00256'175113		MOVL#	3,3,SNC	JEND OF LINK CHAIN?
00257 000765		JMP	THERE	
00260 034423	ENDM:	LDA	3.KEEP	
00261 175400		INC	3,3	JSTEP POINTER IN X DIREC.
00262'010425		ISZ	XCNT	JEND OF X SCAN?
00263'000755		JMP	LOOP	IND
00264'020421		LDA		IYES, GET OLD -VE X COUNT
00265'040422		STA	Ø,XCNT	
00266'020422		LDA	Ø,SIXTN	
00267 034415		LDA	3.NLEFT	
00270'117000		ADD	0,3	31 ROW UP, L.H. SIDE
00271 010415		ISE	YRANG	SEND OF Y SCAN?
00272'000745		JMP	LOOPO	INO
00273 002407		JMP	eSVR3	;YES, EXIT!
00274'000000	XB:	0	and the second second	CALLER ST. I. MARKET C. SAMONO.
	The second se			

00275'000000 YB: 00276'000000 XL: 00277'000000 YL: 00300'000000 YL: 00301'000000 YU: 00302'000000 YU: 00302'000000 SVR 00303'000000 KEE 00304'000000 NLE

XL: 0 YL: 0 XU: 0 YU: 0 SVR3: 0 KEEP: 0 NLEFT: 0

Ø

00305'000000	XRANG:	Ø			C-98
00306.000000	YRANG:	Ø			0.00
00307 '000000	XCNT:	Ø			
00310'000020	SIXTN:	20			
00311'000100	C100:	100			
00312'003000	SV3:	Ø			
00313'000000	BYPAS:	0			
00314'000525'	SVP3R:	SVP3			
00315'000630'	YTGR:	YTGET			
00316'056776	PUSH:	STA	3, esvpai	R	
00317'040013-	A COLUMN TWO IS	STA	Ø,NPNB		
00320 014773		DSE	BYPAS	JONLY COMPUTE CO	S & SIN
00321 000434		JMP	JELLO	J FIRST TIME RC	UND
00001 000 00	; TO GET			IN OF THIS EDGE	
00322'020752		LDA	Ø,XB		
00323'024003-		LDA	1.XA		
00324'122400		SUB	1,0	JXB-XA	
00325'040007-		STA	Ø,COSF	COS SIGN FLAG	
00326'101112		MOVL#	0,0,SZC	J-VE?	
00327'100400		NEG	0.0	YES, GET ABSCAR	-XA)
00330'030014-		LDA	2.L	LENGTH OF EDGE	Dates and all and a
00331'126400		SUB	1.1	recharm of Loop	
00332'142513		SUBL#	States and states	\$XD>=L?	
00333'124001		COM	and the second second	JSET ACI TO 1111	
00334'073101		DIV	171750	,	
00335'101112		MOVL#	0,0,SZC	ROUND UP IF NEC	FSSARY
00336'125400		INC	1,1	51100110 01 11 1120	
00337'044005-		STA	1.005		
00340'020735		LDA	Ø,YB		
00341 024004-		LDA	1.YA		
00342'122400		SUB	1.0	JYB-YA	
00343'040010-		STA	Ø,SINF	ISIN SIGN FLAG	
00344'101112		MOVL#	0.0.SEC		
00345'100400		NEG	0.0		
00346'126400		SUB	1.1		
00347'142513		SUBL#		;YD>=L?	
00350'124001		COM	1,1,SKP	A REAL PROPERTY AND A REAL	
00351'073101		DIV	1717510	7120	
00352'101112		MOVL#	0,0,SEC		
00353'125400		INC	1,1	ROUND UP	
00354'044006-		STA	1.SIN	1100115 01	
00004 044000		UIN			
	IGET TRA	NSFORME	CO-ORDS	S OF X,Y	
			S. There is a second second	+YG*SIN(A)	
	1			-XG*SIN(A)	
	1	the second second			
00355'020013-	JELLO:	LDA	Ø.NPNB	(NP:NB)	
00356'0240105		LDA	1. MSKR		
00357'115300		MOVS	0.3		
00360'123400		AND	1,0	INB IN ACO	
00361'167400		AND	3,1	INP IN AC1	
00362'044535		STA	1.OTHER		
00363'0340015		LDA	3 M1		
00364'117000		ADD	0.3		
00365 031 400		LDA	2.0.3	POINTER TO NEW	BI OCK

00365'031400 00366'0060115 00367 040537 00370'044537 00371 034003-00372'162400

LDA JSR STA STA LDA SUB

Ø.X

2,0,3 JPOINTER TO NEW BLOCK e.PON1 ;GET GLOBAL CO-ORDS 1.Y JACTUAL CONTACT CO-ORDS 3.XA 3,0

-					
	00373 040522		STA	Ø,XG	JREL. TO EDGE START
	00374'034004-		LDA	3.YA	
	00375'166400		SUB	3,1	
	00376'044520		STA	1.YG	
		;			
	00377 006716		JSR	eytgr	
	00400 054524		STA	3.YT	JLOCAL, TRANSFORMED Y
	00401'126520		SUBEL	1+1	
	00402'166512		SUBL#	3,1,SEC	\$15 YT>1?
	00403'002522		JMP	eSVP3	JYES. NOT TOUCHING. EXIT!
	00404'024517		LDA	1. THO	AND A REAL PROPERTY AND A REAL
	00405'137112		ADDL#		115 YT <= - 3?
	00406'002517		JMP	eSVP3	JYES. TOO DEEP. EXIT!
		1			
	00407 .030006-		LDA	2.SIN	JNOW FOR XT
	00410'024506			1.YG	
	00411'102440		SUBO	0,0	
	00412'125112		MOVL#	and the second sec	SET CARRY IF NEG
	00413'124440		NEGO		JAND MAKE AC1 +VE
	00414'073301		MUL	1 1 1 1 1 1 1 1 1 1 1 1 1 1 1 1 1 1 1	
	00415'125112		MOVL#	1,1,SZC	
	00416'101400		INC		JROUND UP
	00417'101002		MOV		JCARRY?
	00420'100400		NEG	and the second s	RESTORE SIGN
	00421'024010-		LDA	1,SINF	
	00422'125102			1,1,520	SIGN OF SIN
	00423'100400		NEG	0.0	CONTRACTOR AND
	00424'115000		MOV	0.3	SHUNT INTO AC3
	00425'024470		LDA	1.XG	CIVER - CONTRACTOR CONTRACTOR
	00426 030005-		LDA	2,005	
	00427 102440		SUBO	0.0	
	00430'125112		MOVL#	1,1,SEC	
	00431'124440		NEGO	1,1	
	00432'073301		MUL		
	00433'125112		MOVL#	1,1,SEC	
	00434'101400		INC	0.0	
	00435'101002		MOV	0,0,SEC	
	00436'100400		NEG	0.0	
	00437 . 024007 -		LDA	1.COSF	
	00440'125102		MOVL	1,1,SEC	
	00441 100400		NEG	0.0	
	00442'117000		ADD	0,3	JADD TO PREVIOUS RESULT
		ILOCAL,	TRANSFOR	RMED X NO	W IN AC3
		3			
	00443 024014-		LDA	1.L	
	00444'166512		SUBL#	3,1,SEC	\$IS XT>L?
	00445'002460		JMP	and the state of the	#YES
	00446'175112		MOVL#	3,3,SEC	\$15 XT<0?
	00447'002456		JMP	eSVP3	\$YES
		JTO FINE	D IF THIS	S CONTACT	ALREADY EXISTS
	00450 0340055		LDA	3. • M5	

00451 0200	11-
00452'1170	00
00453'0544	45
00454'0240	12-
00455'0354	00
00456'1751	12 SEA:
00457 0004	30
00460 0214	00
00461 '0300	105

LDA

ADD

JMP

Ø.NB 0,3 3, PRODL FREMEMBER CONTACT LOCATOR STA LDA 1.NP 3,0,3 LGET POINTER (OR -1) LDA 3,3,SZC MOVL# CLOUD FTHIS CONTACT NOT STORED 0,0,3 #1ST WORD CONTACT LIST LDA 2. MSKR LDA

	-				
- 9	00462'113400		AND	0,2	POINT (EDGE) NUMBER
1	00463'132414		SUB#	1,2,SER	SAME EDGE?
	00464 000405		JMP	WAVES	INO
	00465'021401		LDA	0,1,3	JGET FOINT, BLOCK
	00466'030013-		LDA	Contraction of the second s	COMPOSITE WORD
	00467 112415		SUB#	0,2, SNR	
	00401 112415	AL REA	ADY TOUCH		
	00470.000403	JALALI	JMP	REN	JYES. UPDATE SIN, COS ETC.
1000		WAVES:	LDA	SUPERIO DUN	INO. GET LINK FIELD
	00471'035402	WAVES.	JMP	SEA	THO. OLI LINK FILLD
1	00472'000764		and the second sec	10 To 1 To 10 To 1	RCE TO PREVENT PUNCH-THROUGH
		JADD IN		URMAL FUR	CE TO FREVENT FONGH-TAROUGH
		JIF YT		1 VT	
	00473 024431	REN:	LDA	1,YT	
	00474'125503		INCL	1,1,SNC	
	00475'000466	CHANGE:	JMP	RENEW	ITHIS WORD CAN BE REPLACED
	00476'020405		LDA	Ø,FORCE	一、利益之子、 人名法尔尔 有了不是有
3	00477 025406		LDA	1,6,3	INORMAL FORCE, FN
	00500'107000		ADD	0,1	; ADD IN INCREMENT
1	00501 045406		STA	1,6,3	PUT FN BACK
4	00502 000773		JMP	CHANGE	
11	00503'010000	FORCE:	10000		3 PREVENTIVE FORCE
3	00504'000475'	CHA:	CHANGE		
3	00505'000466		JMP	RENEW-CH	ANGE 1
1	00506'000454		JMP	HEAD-CHA	ANGE 1
		J			
		JNOT	ALREADY	TOUCHING-	THE REAL PROPERTY AND INCOME.
4	00507 024415	CLOUD:	LDA	LAYT	
1	00510'125004		MOV	1,1,SER	;THROW OUT IF
	00511'125112		MOVL#	1,1,SEC	\$ YT>0
1	00512 000554		JMP	WEED	
	00513'002412		JMP	eSVP3	
	00514'020000	FLAG:	20000		
	00515'000000	XG:	Ø		
	00516'000000	YG:	ø		
	00517'000000	OTHER:	ø	CONTACT	F POINT #
	00520.000000	PRODL:	Ø		
	00521'100000	SFLAG:	100000		
	00522 040000	CFLAG:	40000		
	00523'000002	TWO:	2		
	00524'000000	YT:	õ		
	00525 000000	SVP3:	ø		
	00526'000000	X:	Ø	TACTUAL	CONTACT CO-ORDS
	00527 000000	Y:	Ø	JACIOAL	CONTACT CO-ORDS
	00530'000126'	AC2R:	AC2		
			Ø		
1	00531 000000	AC3S:	0		
		TO THE	TOT NEW	ENTRY	A THOMAS
		I INST	LIVE WEW	L.V.I.V.I + + + +	A REAL PROPERTY OF THE PROPERT
-	00532'0340145	ENTER.	LDA	3. FMPT	JGET ADDR. IN EMPT. LOC.
	00533 175112	LIVI LIVI	MOVL#		JIS IT -1?
	00534'000460		JMP	FLOC	JYES. MUST USE MORE CORE
				2,2,3	JGET LINK IN FREE SPACE
	00535'031402		LDA	61610	JOLI LIWA IN THE STALL

2. EMPT JUPDATE EMPTY LOCATOR 00536'0500145 STA 2. PRODL JGET CONTACT LOCATOR LDA 00537'030761 FROG: 00540'021000 0,0,2 LDA 3,0,2 JSTORE NEW ADDR. IN IT 00541 055000 STA 0,2,3 JPUT IN NEW LINK FIELD 00542'041402 STA SNOW PUT IN REST OF DATA ISET ZERO IN FOLLOWING: 00543'102400 0.0 SUB 0,3,3 J S (SHEAR DISP) STA 00544'041403

-	* me				
	00545'041404		STA	0,4,3	; SDEL (INCR. S.D.)
	00546'041405		STA	0,5,3	:NDEL (INCR. N.D.)
	00547'041406		STA	0,6,3	J FN (NORMAL FORCE)
	00550'041407		STA	0,7,3	; FS (SHEAR FORCE)
	00551 054760	HEAD:	STA	3,AC3S	
	00552'024012-	inchio .	LDA	LINP	
	00553'032755		LDA	2, PAC2R	
	00554'0060175		JSR	e.TYP	
	00555'101300		MOVS	0,0	
	00556'107000			110000000000	
	00557 034752		ADD	0,1	
	00560 045400		LDA	3,AC35	WEAD OF LIST
			STA	1,0,3	THEAD OF LIST
	00561 020013-		LDA	Ø,NPNB	
	00562'041401	DEMELLA	STA	0,1,3	12ND WORD
	00563 020743	RENEW:	LDA	0.X	
	00564'041412		STA	0,12,3	JGLOBAL X OF CONTACT
	00565'020742		LDA	0,Y	
	00566'041413		STA	0,13,3	JGLOBAL Y OF CONTACT
	00567 020006-		LDA	Ø,SIN	
	00570'041410		STA	0,10,3	JSIN
	00571 020005-		LDA	Ø,COS	
	00572'041411		STA	0,11,3	JCOS
	00573'020721		LDA	Ø.FLAG	J"PRESERVE" FLAG
	00574'030010-		LDA	2, SINF	
	00575'151113		MOVL#	2,2, SNC	
	00576'000403		JMP	.+3	
	00577 024722		LDA	1,SFLAG	
	00600'123000		ADD	1,0	JADD IN SIN FLAG IF -VE
	00601 '030007-		LDA	2,COSF	
	00602'151113		MOVL#	2,2, SNC	
	00603'000403		JMP	.+3	
	00604'024716		LDA	1.CFLAG	
	00605'123000		ADD		ADD IN COS FLAG IF -VE
	00606'025400		LDA	1,0,3	JOLD HEAD
	00607 030420		LDA	2. SCMSK	FOLD HERD
	00610'147400		AND	2,1	
	00611 107000		ADD	0,1	
			STA		INEW HEAD
	00612'045400		JMP		JNEW HEAD
	00613'002712		Contraction of the second s	eSVP3	NEXT EDEC LOCATION
	00614'0340075	FLUC:	LDA		INEXT FREE LOCATION
	00615'0200205		LDA	and the second se	MAX. ADDRESS POSSIBLE
	00616'024015\$		LDA	1. PSIZ	
	00617'167000		ADD	3,1	
	00620'122513		SUBL#		STORAGE OVERFLOW?
	00621'000404		JMP	and the second second second	JNO, OK
	00622'0060135		JSR	e.PRN1	JYES, RING THE BELL
	00623'000007		7		
	00624'002701		JMP		JEXIT WITHOUT STORING
	00625'0440075	NOG:	STA	1 . • M7	JUPDATE FREE POINTER
	00626'000711		JMP	FROG	
	00/07/017777	SCMSV .	17777	TO MASH	OFF OLD S.C.P FLAGS

00627 017777 SCMSK: 17777 JTO MASK OFF OLD S,C,P FLAGS

; TO CALCULATE YT J INPUT: YG IN AC1 YTGET: STA 3,YTSAV 00630'054435 LDA 2,005 00631 030005-SUBO 0.0 00632'102440 MOVL# 1,1,SZC 00633'125112 1 . 1 NEGO 00634'124440 00635'073301 MUL

	00636'125112		MOVL#	1,1,SEC	
	00637'101400		INC	0.0	
	00640'101002		MOV	0,0,SZC	
	00641 100400		NEG	0.0	
	00642'024007-		LDA	1,COSF	
	00643'125102		MOVL	1,1,SEC	
	00644'100400		NEG	0,0	
	00645'115000		MOV	0.3	JPARTIAL SUM IN AC3
	09646'024647		LDA	1,XG	
	00647 030006-		LDA	2.SIN	
	00650'102440		SUBO	0.0	
	00651'125112		MOVL#	1,1,SZC	
	00652'124440		NEGO	1 + 1	
	00653'073301		MUL		
	00654'125112		MOVL#	1,1,SZC	
	00655'101400		INC	0.0	
	00656'101002		MOV	0,0,SZC	
	00657'100400		NEG	0.0	
	00660 024010-		LDA	1,SINF	
	00661'125102		MOVL	1,1,SZC	
	00662'100400		NEG	0.0	
	00663'116400		SUB	0,3	SUBTRACT FROM PREVIOUS RESULT
	00664'002401		JMP	eytsav	
	00665'000000	YTSAV:	Ø		
	00666'024631	WEED:	LDA	1.OTHER	JCONTACT CANDIDATE
		ROUTIN	E TO WEEL	D OUT IM	POSSIBLE CONTACTS
	00667 044444		STA	1,SWIT	
	00670'125005		MOV	1,1, SNR	; ZERO?
	00671 000404		JMP	TOAD	JYES
	00672'102520		SUBZL	0.0	
	00673'106400		SUB	0,1	JTRY [POINT-1]
	00674'000402		JMP	GETIT	
	00675'126520	TOAD:	SUBEL	1,1	JTRY POINT #1
	00676'0060125	GETIT:	JSR	e.PON2	; (PONT ALREADY PRIMED)
	00677'050435		STA	2, SV2	
	00700 034003-		LDA	3.XA	
	00701 162400		SUB	3,0	
	00702'040613		STA	Ø,XG	REL X
	00703'034004-		LDA	3.YA	
	00704'166400		SUB	3,1	FREL Y
	00705'004723		JSR	YTGET	
	00706'024615		LDA	1.TWO	
	00707'167112		ADDL#	3,1,SZC	:YT1<=-2?
	00710'002615		JMP	esvp3	#YES. IMPOSSIBLE CONTACT
	00711'020422		LDA	Ø,SWIT	
	00712'101112		MOVL#	0,0,SZC	#2ND TIME ROUND
	00713 000617		JMP	ENTER	SYES. STORE THE CONTCT
	00714'030420		LDA	2, SV2	
	00715 025000		LDA	1,0,2	#CONTROL WORD
	00716'0340105		LDA	3. MSKR	
	00717 167400		AND	3,1	INO. OF POINTS (PMAX)
	00720'176000		ADC	3,3	2-1 contraction and an and a second s
	00721 054412		STA	3.SWIT	ESET FOR FXIT 2ND TIME

00721'054412 00722'101004 00723'000403 00724'167000 00725'000751 00726'101400 NEWT: 00727'106415 00730'102400

STA MOV JMP ADD JMP INC SUB# SUB 3,SWIT #SET FOR EXIT 2ND TIME 0,0,SZR NEWT #SWIT MUST BE >0 3,1 #TRY POINT (PMAX-1) GETIT 0,0 #OTHER +1 0,1,SNR #IS IT EQUAL TO PMAX? 0,0 #YES. USE POINT #0

Ø . END

SV2:

C-103

			DUITS CANCE

		.TITL	REBOX	
	JTO RE-	CLASSIFY	(IF NEC	ESSARY) ALL
	THE PO	INTS OF	ONE BLOCK	K IN NEW
	JBOXES.			
	1			
	i	JSR 0.R	EBX	
	I CIN	And the second s	Contraction ()	ER TO BLOCK DATA,
	1			ER TO LOCATOR)
	:402 15	PRESERV		
	2000 10	.ENT		JTEMP TEST ENTRY
		.ENT		REBZ, MSKR
		and the second se		MA PON1 PON2 PRES LENG
		.ZREL		
00000-0000000.	.REBX:	REBX		
00001-000002'		REBE	IENTRY	WITH NB IN AC1
00002-000377	.MSKR:	377	201101	
00002 000011	•HJAN	.NREL		
00000.0200015	REBX:	LDA	Ø M1	
00001'106400	acon.	SUB	0.1	
00002'044506	REBE:	STA		JREGENERATE NB
00003'054477	neber	STA	3, SVRB3	
00004'050475		STA	2,5V2	
00005'021000		LDA	0,0,2	
00006'024002-		LDA	1. MSKR	
00007 123400		AND	1,0	
00010 040504		STA	Ø,PCNT	
00011'126400		SUB	1,1	
00012'044475		STA	1.NP	
00013'0060045		JSR	0.PONI	
00014'000403		JMP	PLACE	
00015'024472	COW:	LDA	1.NP	
00016'0060055	00.	JSR	PON2	
00017'176520	PLACE:	SUBEL	3,3	JCHECK IF ON SCREEN
00020'162512	LHOL.	SUBL#	3,0,SEC	
00021 000523		JMP	FIX	JYES, FIX THE BLOCK
00022'166512		SUBL#	3,1,SEC	
00023'000521		JMP	FIX	
00024'034466		LDA	3,01777	
00025'162513		SUBL#		; X>=1023 (DECIMAL)?
00026'000516		JMP	FIX	TAP-1025 (DEGINAL).
00027 034464		LDA	3,01414	
00030'166513		SUBL#		1Y>=780 (DEC)?
00031 000513		JMP	FIX	11100 (0207)
00032*044453		STA	LINY	
00002 044400	;	214		
00033'105000	CONT:	MOV	0,1	FIND NEW BOX
00034'0340025	00111.	LDA	3. • M3	TIND NEW DOX
00035'030447		LDA	2,0100	
00036'102400		SUB	0,0	
00037 073101		DIV	070	
00040'137000		ADD	1,3	
00010 101000		100		

SUB LDA DIV ADDZL ADDZL ADD STA MOV LDA 0.0

1.NY

1,1 1,1 1,3 JBOX ADDR. IN AC3 3,BOX 3,2 Ø,NB

				0-105
00052'024435		LDA	1.NP	
00053'125300		MOVS	1,1	
00054'123000		ADD	1.0	J(NP:NB) IN ACØ
00055 004502		JSR	FIND	JFIND OLD BOX
00056'000461		JMP	ITER	JSUCCESS! NO CHANGE
00057 034437		LDA	3,LIST	FAILURE! MUST SEARCH AROUND
00060'054426	WINE:	STA	3, POINT	
00061 030430		LDA	2,80X	
00062'025400		LDA	1,0,3	
00063'125005		MOV	1,1, SNR	
00064'000453		JMP	ITER	WHERE IS IT
00065'133000		ADD	1,2	FUNCTION IN
00066'0240025		LDA	1M3	
00067'132512		SUBL#	1,2,SEC	
00070'000406		JMP	NEXT	INON-EXISTENT BOX
00071 0240035		LDA	1M4	MON-ENISTENT BOX
00072'132513		SUBL#	1,2, SNC	
		Contraction of the second s	the second second second second	; DITTO
00073'000403		JMP	NEXT	J DITTO JTRY THIS BOX
00074'004463		JSR	FIND	
00075'000433	NEVT.	JMP	FOUND	FOUND IT!
00076'034410	NEXT:	LDA	3, POINT	INO GOOD. TRY NEXT BOX
00077'175400		INC	3,3	
00100'000760	cuo.	JMP	WINE	
00101.000000	SV2:	0		
00102'000000	SVRB3:	Ø		
00103'000000	OLD:	0		
00104'000100	C100:	100		
00105.000000	NY:	Ø		
00106'000000	POINT:	0		
00107 000000	NP:	0		
00110'000000	NB:	Ø		
00111 000000	BOX:	Ø		
	C1777:	1777		
00113'001414	C1414:	1414		
00114'000000	PCNT:	Ø		
00115'004000	FBIT:	4000	IMASTER	FIX BIT (OVERRIDES MAN. BIT)
00116 000117	LIST:	•+1		
				ES, IN EXPECTED
	JORDER (BLE OCCUR	ANCE
00117 000020		20		
00120'177777		-1		
00121'000001		1		
00122'177760		-20		
00123'000017		17		
00124'000021		21		
00125'177757		-21		
00126'177761		-17		
00127.000000	1200120000000000	Ø		
00130'034753	FOUND:	LDA	3,0LD	JGET CALLING ADDR
00131 025001		LDA	1,1,2	JEXISTING LINK
00132'045400		STA	1,0,3	BRIDGE ACROSS ENTRY
			10 Mar 10 Mar	

00133'034756 00134'021400 00135'051400 00136'041001 00137'010750 ITER: 00140'014754 00141'000654 00142'030737 00143'000430 LDA LDA STA STA ISZ JMP LDA JMP 3,80X 0,0,3 2,0,3 0,1,2 NP PCNT COW 2,5V2 PUP INEW BOX ADDRESS
IPOINTER (OR -1)
IPUT IN NEW ADDRESS
ICOMPLETE LINK
INEXT POINT
INEXT POINT IF NOT DONE
IUPDATE ANY PRESS. SEGS

1 .NY 00144'044741 FIX: STA LDA 1,0,2 00145'025000 00146'034747 LDA 3,FBIT 3,1, SNR ; SKIP IF FLAG ALREADY SET AND# 00147'167415 00150'167000 ADD JADD IN MASTER FIX FLAG 3,1 ; PUT CONTROL WORD BACK 00151 045000 STA 1,0,2 JALLOW "INVISIBLE" SUB 3,3 00152 176400 ; BLOCKS 00153 055020 STA 3,20,2 3 TO 00154 055021 STA 3,21,2 ; INTERACT 3,22,2 00155'055022 STA 00156'000655 JMP CONT ;KEEP GOING ; ROUTINE TO FOLLOW CHAIN TO FIND (NP:NB) STA 2,0LD ;CALLING ADDR 00157 050724 FIND: JADDR OF 1ST WORD 00160 031000 LDA 2,0,2 MID JMP 00161 030407 LDA 1,0,2 00162 025000 ROUND: 0,1, SNR ; COMPARE 00163'106415 SUB# JMP SUCCESS! ADDR. IN AC2 00164'001400 0,3 INC 2,1 00165'145400 00166'044715 STA 1,OLD JOLD LINK ADDR. JGET LINK LDA 2,1,2 00167'031001 MOVL# 2,2,SEC JEND OF CHAIN? 00170'151112 MID: JYES. FAILURE EXIT JMP 1,3 00171 001401 00172'000770 JMP ROUND ; ; ROUTINE TO UPDATE FX, FY IN ANY ; PRESSURE SEGMENT FOR BLOCK NB ; 00173'021000 PUP: LDA 0,0,2 00174'024506 LDA 1, PMSK 1,0,SNR JOUICK CHECK FOR PRESS. 00175'123415 AND# 00176'002704 JMP €SVRB3 INONE FOR THIS BLOCK 00177'0300065 LDA 2, PRES 00200 034710 GRAPE: LDA 3,NB 00201 151113 PLUM: MOVL# 2,2, SNC 00202'000403 JMP .+3 00203 030676 LDA 2, SV2 00204 002676 JMP eSVRB3 JEND OF PR. SEG. LIST 00205 025000 LDA 1,0,2 INPNB THIS SEG. 00206.050005-LDA Ø, MSKR 00207 123400 AND 1,0 JNB1 (BLOCK #) 00210'116415 SUB# 0,3, SNR ; SAME BLOCK? JYES; UPDATE FX, FY 00211'000403 JMP PRUNE INO, GET NEXT LINK 00212 031002 LDA 2,2,2 PLUM 00213'000766 JMP SUBS JNP1 (EDGE #) 00214'106700 PRUNE: 0,1 00215 050466 STA 2, PR2 JCURRENT PR. LIST POINTER 00216 035001 LDA 3,1,2 **JFORCE** 00217 054465 STA 3,FORCE 00220 044465 1,NPREM ; REMEMBER 1ST CORNER STA

201-3

00222'117000 00223'031400 00224'0060075 00225'040461 00226'0060045 00227'040460 00230'044460 00230'044460 00231'024454 00232'125400

00221 0340015

ADD LDA JSR STA JSR STA STA LDA INC

LDA

0,3 2,0,3 31 0.LENG 30 0.L 0.XA 1.YA 1.NPREM

3, .M1

1 . 1

JBLOCK POINTER JGET LENGTH

0.000

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00233'021000		LDA	0,0,2	
00234'034002-		LDA	3, MSKR	
00235'163400		AND	3,0	INC
00236'106415		SUB#	0,1, SNR	JCHECK FOR LAST CORNER
00237'126400		SUB	1 . 1	
00240'0060055		JSR	e.PON2	
00241 030446		LDA	2,XA	
00242'112400		SUB	0,2	(XA-XB)
00243'155000		MOV	2,3	JSAVE FOR SIGN
00244'044445		STA	1,YB	Forthe Fort Digit
00245'024437		LDA	1.FORCE	
00246'102440		SUBO	0.0	
00247'151112		MOVL#		JCHECK SIGN
00250'150400		NEG	2,2	VOILON DION
00251 073301		MUL	636	
00252'030434		LDA	2,1	
00252 030434		and the second se	CIL	
00254 175112		DIV	2 2 526	DESTORE STON
		MOVL#	the first the second second	RESTORE SIGN
00255'124400		NEG	1,1	
00256'044434		STA	1.FY	
00257 030432		LDA	2,YB	
00260'020430		LDA	Ø,YA	
00261'112400		SUB	0,2	J (YB-YA)
00262'155000		MOV	2,3	
00263'024421		LDA	1,FORCE	
00264'102440		SUBO	0,0	
00265'151112		MOVL#	2,2,520	
00266'150400		NEG	5,2	
00267 073301		MUL		
00270'030416		LDA	2,1	
00271 073101		DIV		; (YB-YA)*F/L
00272'175112		MOVL#	3,3,SZC	
00273124400		NEG	1 > 1	JFX
00274'030407		LDA	2, PR2	
00275'045004		STA	1,4,2	ISTORE FX IN LIST
00276'024414		LDA	1.FY	
00277'045005		STA	1,5,2	JFY IN LIST
00300 031002		LDA	2,2,2	JLINK
00301 '000677		JMP	GRAPE	
00302 000400	PMSK:	400		
00303 .000000	PR2:	Ø		
00304.000000	FORCE:	Ø		
00305'000000	NPREM:	Ø		
00306'000000	L:	Ø		
00307 '000000	XA:	Ø		
00310.000000	YA:	0		
00311'000000	YB:	Ø		
00312'000000	FY:	ø		
	1 1 1 1 1 1 1 1 1 1 1 1 1 1 1 1 1 1 1	FND		

100					
			.TITL	MOTIO	
		IROUTIN	E TO APP	LY LAW OF	F MOTION TO ALL BLOCKS
			. ENT	.MOT .R	DT. TREC
			•EXTD	.M1DI	SB, REBX, PFLG
			. EREL		
	00000-000001*	.MOT:	MOT		
	00001-000140	.ROT:	140		
	00002-000040	.TREC:	40	11/TDEL	
			.NREL		
	00000.000000	SAVE:	Ø		
	00001 '054777	MOT:	STA	3, SAVE	
	00002 0340015		LDA	3 M1	
	00003 054547	MOT1:	STA	3. BLOCK	
	00004'031400		LDA	2,0,3	
	00005'151005		MOV	2,2, SNR	
	00006 002772		JMP	e SAVE	JEXIT!
	00007 021014		LDA	0,14,2	JAREA
	00010'101005		MOV	0,0,SNR	
	00011 000524		JMP	SKIP	JZERO AREA. SKIP!
	00012.051000		LDA	0,0,2	
	00013'024540		LDA	1. FMSK	JTO DETECT "FIXED" FLAG
	00014'107404		AND	Ø,1,SZR	
	00015'000520		JMP	SKIP	
	00016'021007		LDA	0,7,2	FXSUM
	00017 025005		LDA	1,5,2	JOLD X-VEL
	00020'004535		JSR	ADDMX	The second secon
	00021 045005		STA	1,5,2	JNEW X-VEL
	00022 050532		STA	2, SV2	
	00023'030002-		LDA	2. TREC	
	00024'102400		SUB	0.0	
	00025'135000		MOV	1,3	KEEP FOR SIGN
	00026'125112		MOVL#	1,1,SEC	TROVE FOR IVILLI SECTION TONS
	00027 124400		NEG	1.1	
	00030'146512		SUBL#		BYPASS IF ANSWER WILL BE Ø
	00031 000516		JMP	FLIP	
	00032 073101		DIV		; INTEGER DIVIDE
	00033'030521		LDA	2,5V2	
	00034'021002		LDA	0,2,2	; XC(LOW)
	00035'175112		MOVL#	3,3,SEC	
	00036'000405		JMP	FLIT	I WAS NEGATIVE
	00037'123023		ADDZ	1,0, SNC	
	00040'000417		JMP	OK	
	00041'011001		ISZ	1,2	JINCREMENT XC(HIGH)
	00042.000405		JMP	CHECK	
	00043'124400	FLIT:	NEG	1 . 1	
	00044'123022		ADDZ	1,0,SEC	
	00045'000412		JMP	OK	
	00046'015001		DSZ	1,2	JDECREMENT XC(HIGH)
	00047 045020	CHECK:	STA	1,20,2	
	00050'041002		STA	0,2,2	
	00051 '024501		LDA	1,BLOCK	
	00052'0060035		JSR	C.REBX	;RE-CLASSIFY THIS BLOCK
	00053'0340045		LDA	3. PFLG	
	00054'175005		MOV	3,3, SNR	
	00055'0060025		JSR	0.DISB	
	00056'000403		JMP	NUT	
	00057'045020	OK:	STA	1,20,2	J DELTA-XC
	00060'041002		STA	0,2,2	JNEW XC(LOW)
		3			
	00061'021016	NUT:	LDA	0,16,2	FISUM

-					
	00062'025015		LDA	1,15,2	JOLD Y-VEL
	00063'004472		JSR	ADDMX	
	00064'045015		STA	1,15,2	INEW Y-VEL
	00065'030002-		LDA	2. TREC	
	00066'102400		SUB	0,0	JCLEAR HI PART
	00067 135000		MOV	1,3	JSAVE FOR SIGN
	00070'125112		MOVL#	1,1,SEC	
	00071'124400		NEG	1,1	
	00072'146512		SUBL#	2,1,SEC	BYPASS IF ANSWER WILL BE Ø
	00073'009451		JMP	FLOP	
	00074'073101		DIV	JINTEGER	DIVIDE
	00075'030457		LDA	2, SV2	
	00076'021004		LDA	0,4,2	JYC(LOW)
	00077'175112		MOVL#	3,3,SEC	
	00100'000405		JMP	FLITS	
	00101'123023		ADDE	1,0, SNC	
	00102'000417		JMP	OKS	
	00103 011003		ISZ	3,2	JINCREMENT YC (HIGH)
	00104'000405		JMP	CHECS	
	00105'124400	FLITS:	NEG	1 = 1	
	00106'123022		ADDZ	1,0,SZC	
	00107 009412		JMP	OKS	
	00110'015003		DSZ	3,2	JDECREMENT YC(HIGH)
	00111'045021	CHECS:	STA	1,21,2	
	00112'041004		STA	0,4,2	
	00113'024437		LDA	1, BLOCK	
	00114'0060035		JSR	e.REBX	JRE-CLASSIFY
	00115'034004\$		LDA	3. PFLG	
	00116'175005		MOV	3,3, SNR	
	00117'0060025		JSR	e.DISB	PLOT JUST THIS BLOCK
	00120'000460		JMP	CLOT	
	00121'045021	OKS:	STA	1,21,2	JDELTA-YC
	00122'041004		STA	0,4,2	JNEW YC(LOW)
		3			
	00123'000455		JMP	CLOT	;NOW FOR MOMENTS
		3			124 A 4 4 4 4 4 4 4 4 4 4 4 4 4 4 4 4 4 4
	00124'021023	CLOT1:	LDA	0,23,2	;X LOAD
	00125'041007		STA	8,7,2	;INIT . XFSUM
	00126'021024		LDA	0,24,2	JY LOAD
	00127 025014		LDA	1,14,2	GRAVITY FORCE
	00130'122400		SUB	1,0	
	00131 041016		STA	0,16,2	JINIT. YFSUM
	00132'102400		SUB	0,0	
	00133'041017		STA	0,17,2	JSET MSUM TO Ø
	00134'000405		JMP	PAST	
	00135'102400	SKIP:	SUB	0.0	
	00136'041007		STA	0,7,2	JXFSUM=0
	00137'041016		STA	0,16,2	;YFSUM=0
	00140'041017		STA	0,17,2	INSUM=0
			104	0 0 000	

3,BLOCK 00141'034411 PAST: LDA 00142'175400 INC 3,3 MOT1 JMP 00143'000640 00144'030410 2, SV2 FLOP: LDA SET DELTA-YC TO Ø 0,21,2 STA 00145'041021 00146'000432 CLOT JMP 00147 030405 2, SV2 FLIP: LDA 0,20,2 00150'041020 STA 00151'000710 JMP NUT 00152 000000 BLOCK: Ø J"FIXED" NASK 00153'014000 FMSK: 14000

÷					
	00154'000000	SV2:	Ø		
		3			
					AN UPPER
				THE ANSWER	
	00155'125020	ADDMX:	MOVE	1.1	JCLEAR CARRY
	00156'125112		MOVL#	1,1,SZC	
	00157 000405		JMP	A1	
	00160'101113		MOVL#	0,0,SNC	POTU AND
	00161 000407		JMP	POS	JBOTH +VE
	00162'107000	DIF:	ADD	0,1	JBOTH SIGNS DIFFERENT
	00163'001400		JMP	0,3	JEXIT
	00164'101113	A1:	MOVL#	0.0.SNC	
	00165'000775		JMP	DIF	BOTH DIF
	00166'124400		NEG	1,1	BOTH -VE
	00167 100440		NEGO	0.0	INEGATE BOTH. SET CARRY
	00170'107000	POS:	ADD	0.1	
	00171'020406		LDA	Ø,MAX	PTUR TOOLSI'IG IS
	00172 106432		SUB2#	The state of the s	LIMIT MAX VELOCITY
	00173'105000		MOV	0,1	
	00174'125002		MOV	1,1,SZC	
	00175 124400		NEG	1,1	JYES, NEGATE!
	00176'001400		JMP	0,3	JEXIT
	00177 037777	MAX:	37777		
	00200'126400	CLOT:	SUB	1,1	JCLEAR LOWER
	00201 021017		LDA	0,17,2	JMSUM
	00202'031013		LDA	2,13,2	JIAY2 BERGERICTI
	00203'115000		MOV	0,3	JSAVE M FOR LATER
	00204'101112		MOVL#	0,0,SEC	
	00205'100400		NEG	0.0	; ABS(MSUM)
	00206'142432		SUB2#	2,0,SZC	CHECK FOR OVERFLOW
	00207'124001		COM	1,1,SKP	
	00210'073101		DIV		
	00211'125220		MOVER	1,1	J) .ROT ERR
	00212'125220		MOVER	1,1	;)/8
	00213'125220		MOVER	1 > 1	3)
	00214'175102		MOVL	3,3,SEC	
	00215'124400		NEG	1,1	RESTORE SIGN
	00216'121000		MOV	1.0	
	00217 030735		LDA	2,5V2	
	00220 025006		LDA	1,6,2	JOLD ALPHA-DOT
	00221 004734		JSR	ADDMX	
	00222'045006		STA	1,6,2	INEW ALPHA-DOT
	00223'030001-		LDA	2,.ROT	
	00224'102400		SUB	0.0	
	00225'135000		MOV	1,3	
	00226'125112		MOVL#	1,1,SZC	
	00227 124400		NEG	1+1	
	00230'146513		SUBL#	2,1, SNC	ICHECK FOR UNDERFLOW
	00231 000410		JMP	TREE	
	00232 030722		LDA	2, SV2	
	000001041000		CTA.	0.00 0	. 2000 DELTA-ALDUA

00234'000670	
00235 024715	CLOT2
00236'006003\$	
00237 000665	
00240'040000	TEST:
00241 073101	TREE:
00242'030712	
00243'175102	
00244'124400	

00233'041022

JMP LDA JSR JMP 40000 DIV LDA MOVL NEG

STA

CLOT1 1,BLOCK 0.REBX CLOT1

2, SV2

3,3,SEC 1,1

0,22,2

JNO MORE TO DO

JZERO DELTA-ALPHA

00245'021012		LDA	0,12,2	JALPHA (OLD)
00246'123000		ADD	1,0	JADD IN D-ALPHA
00247 125120		MOVEL	1.1	JMAKE UP TOTAL SHIFT
00250'125120		MOVEL	1.1	; TO 8 BITS
00251'125120		MOVEL	1.1	, 10 0 BIIS
00252'045022				ADDI TA ALDUA
		STA	1,22,2	
00253'040514		STA	Ø,SIGN	
00254'105102		MOVL		J-VE? (GARBAGE IN ACI)
00255'100400		NEG	0.0	;YES (C IS SET)
00256'024762		LDA	I,TEST	
00257'122513		SUBL#	1.0.SNC	; IS ALPH>= 1/64?
00260'000405		JMP	CHAN	JYES. INCR. COS & SIN
00261 101002		MOV	0,0,SEC	JWAS SIGN -VE?
00262 100400		NEG	0.0	JYES. RESTORE IT
00263 041012		STA	0,12,2	JALPHA (NEW)
00264'000640		JMP	CLOT1	
00265'122462	CHAN:	SUBC	A DESCRIPTION OF THE ADDRESS	SUBTRACT ALPH(MAX)
00266'100400		NEG	0,0	JOODTINGT ALL MUNAN
00267'041012		STA		JALPHA (NEW)
00270 024500		LDA	1, AMAX	JALFRANKENJ
00271 031011				+ CTN
		LDA	2,11,2	1214
00272'102400		SUB	0.0	
90273'073301		MUL		JMULT. BY AMAX (1/64)
00274'125112		MOVL#	1,1,SZC	
00275'101400		INC	0.0	JROUND UP
00276'030656		LDA	2,5V2	(SIN*AMAX NOW IN CAØ)
00277 025000		LDA	1,0,2	JSIN FLAG
00300'044471		STA	1.SFLAG	
00301 125100		MOVL	1 = 1	JPUT FLAG IN CARRY
00302 034465		LDA	3,SIGN	JD(ALPHA) FLAG
00303'175112		MOVL#	3,3,SZC	
00304'175060		MOVC	3,3	
				;1S COS FLAG SET?
00306'125060				:YES. COMP. CARRY
00307'035010				
				SAME SIGNS, C & D(C)?
00311'000404				JYES. SUBTRACT!
00312'117022				
00313'176000				JSET TO MAX IF OVERFLOW
and the second se				Juli to that it over bon
00314'000413		JMP		(200) 0-200.
00315'116422	CARO:			1005-0(005)
00316'000411		JMP		
00317 174400				
00320 025000		LDA		
00321*125100		MOVL		
00322'125100		MOVL		
00323'125060		MOVC	1 > 1	JODMPLEMENT COS FLAG
000041105000		MOUP	1 . 1	

00327'025010	PRUNE:	LDA	1,10,2	JOLD COS	
00330'055010		STA	3,10,2	SNEW COS	
00331 030437		LDA	2.AMAX		
00332'102400		SUB	0.0		
00333'073301		MUL			
00334'125112		MOVL#	1,1,SEC		
00335'101400		INC	0.0	; RIUND UP	
00336'024433		LDA	1,SFLAG		
00337'125100		MOVL	1 > 1	JEECOMES COS	FLAG
00340'125100		MOVL	1,1	JOOW IN CARY	

1 . 1

1 . 1

1,0,2

JLPDATE CONTROL WORD

MOVR

MOVR

STA

00324'125200 00325'125200

00326 045000

-					
	00341'034426		LDA	3,SIGN	JD(ALPHA) FLAG
	00342'175112		MOVL#	3,3,SEC	
	00343'175060		MOVC	3,3	
	00344'030610		LDA	2,5V2	
	00345'025000		LDA	1,0,2	JNEW CONTROL WORD
	00346'125112		MOVL#	1,1,SEC	JIS SIN FLAG SET?
	00347'125060		MOVC	1 = 1	JYES. COMPLEMENT C
	00350'035011		LDA	3,11,2	;OLD SIN
	00351'125002		MOV	1,1,SZC	SAME SIGNS, S & D(S) ?
	00352'000404		JMP	SARO	JNO. SUBTRACT!
	00353'117022		ADDZ	0,3,52C	;SIN+D(SIN)
	00354'176000		ADC	3,3	JOVERFLOW
	00355'000410		JMP	PLUM	
	00356'116422	SARO:	SUBE	0,3,SZC	SIN - D(SIN)
	00357 000406		JMP	PLUM	INO SIGN CHANGE
	00360'174400		NEG	3,3	
	00361'125100		MOVL	1+1	
	00362'125060		MOVC	1.1	COMPLEMENT SIN FLAG
	00363'125200		MOVR	1 + 1	
	00364'045009		STA	1,0,2	JUPDATE CONTROL WORD
	00365'055011	PLUM:	STA	3,11,2	JNEW SIN
	00366'000647		JMP	CLOT2	JROTATION DONE
	00367 '000000	SIGN:	Ø		
	00370.001000	AMAX:	1000	11/128 ((DEC)
	00371 '000000	SFLAG:	0		
			. END		

. END

.TITL DISPL ITO DISPLAY ALL BLOCKS, CENTROIDS ON J THE SCREEN, OR ON PAPER 3 JSR @.DISS 3 SCREEN ENTRY ... ; PAPER ENTRY 3 JSR @.DISP . . . 3 JSR @.DISB PLOT SINGLE BLOCK 13 . . . ON THE SCREEN 3 (AC2: BLOCK POINTER) 3 3 ; JSR @.LPLS TO PLOT LOAD VECTORS ... ON SCREEN 3 3 .DISS, .DISP, .DISB, .NVEC, .LPLS .ENT ·PLTS, RLNC, PON1, PON2, M1, PRN1 .EXTD .MSKR, .NUM, .SCAL, .LFAP, .LENG •EXTD . IPRN, .MESS, .ALPH, .UD, .AXIS .EXTD .EXTD .PRES, . IPRN, .NVEC .EXTN FEET . ZREL 00000-000000 .PLOT: Ø 00001-000100' DISS .DISS: .DISP: DISP 00002-000056' SINGLE BLOCK ENTRY DISB 00003-000053' .DISB: .LPLS: LPLS 00004-000271* 00005-000000 .NVEC: JFLAG TO PRINT LOADS 0 .NREL DRIVE: 00000'000001 1 . RDX 10 000012 ITO PLOT AXES AXES: STA 3, AXSAV 00001 054444 LDA 0,A1 00002 020444 00003'024444 LDA 1,A2 JSR e.PLTS 00004'0060015 Ø 00005 000000 JSR e.ALPH 00006'0060165 LDA Ø,.UD 00007 0200175 00010'101005 MOV 0,0, SNR JMP PAXSAV 00011 002434 00012'0060145 JSR e.IPRN 00013'000004 4 JSR 00014'0060155 e.MESS 00015'177777 FEET 59 00016'000073 750 00017 001356 LDA 0,A3 00020'020430 LDA 1,A4 00021 024430 JSR e.PLTS 00022 0060015 00053,000000 0 e.ALPH JSR 0,.UD LDA JSR e.IPRN 4 JSR e.MESS FEET 781 35 e.AXIS JSR

00024 0060165 00025'0200175 00026 0060145 00027 '000004 00030 0060155 00031 '000015' 00032'001415 00033'000043 00034'0060205

-					
	00035'001412		778		
	00036'000001		1		
	00037 000001		1		
	00043'0060205		JSR	8.AXIS	
	00041 176366		-778		
	00042'000001		1		
	00043'000001		1		
	00044'002401		JMP	BAXSAV	
	00045'000000	AXSAV:	Ø		
	00046'000003	A1:	3		
	00047 001356	A2:	750		
	00050.001265	A3:	693		
	00051 000043	A4:	35		
	000010		.RDX	8	
	000010		*npn	0	
	00052'000273'	DIR:	DIREC		
	00053 0200015	DISB:	LDA	Ø. PLTS	
	00054'040000-	0156.	STA	Ø. PLOT	
	00055'000465		JMP	SING	
	00056'054524	DISP:	STA	3,5V3	
	00057 020721	TRY:	LDA	Ø,DRIVE	
	00060'062074		DOB	ØLINC	
	00061'020460		LDA	Ø,BLK	
	00062'024455		LDA	1,NBLK	
	00063'030455		LDA	2.CORE	
	00064 050000-		STA	2. PLOT	APEND IN DADED DI OF DOUTINE
	00065'0060025		JSR	e.RLNC	JREAD IN PAPER PLOT ROUTINE
	00066'125005		MOV	1,1,5NR	
	00067 000403		JMP	.+3	
	00070'063077		HALT		; TAPE ERROR
	00071 000766		JMP	TRY	
	00072'020444		LDA	Ø,FFP	
	00073'040441		STA	ØJFFR	
	00074'0200125		LDA	Ø. LPAP	ILOADS NEEDED?
	00075'101004		MOV	0,0,SZR	
	00076'006754		JSR	edir	JYES
	00077'000407		JMP	SUN	
	00100'0200015	DISS:	LDA	Ø. PLTS	
	00101 040000-		STA		SCREEN-PLOT POINTER
	00102'020433		LDA	Ø,FFS	
	00103'040431		STA	Ø,FFR	
	00104'054476		STA	3, SV3	
	00105'004674		JSR	AXES	PLOT AXES ON SCREEN ONLY
	00106'034005\$	SUN:	LDA	3 M1	
	00107 054472	RAIN:	STA	3, BPNT	
	00110'031400		LDA	2,0,3	
	00111'151005		MOV	2,2, SNR	
	00112'000414		JMP	FINAL	INO MORE BLOCKS
	00113'021014		LDA	0,14,2	JAREA
	00114'101005		MOV	0,0,SNR	JZERO?
	0011151000101		B + 4 175	1.7.8.5	NEC CUID THEC DI GOL

JMP JYES, SKIP THIS BLOCK 00115'000406 WIND 00116'021000 LDA 0,0,2 TINGSTRONG. 00117 024505 LDA 1,FMSK 1.0.SER JFIXED BLOCK? AND# 00120'123414 JYES, PRINT AN "F" 00121'006413 JSR effr 00122.004420 JPLOT THIS BLOCK JSR SING LDA 3, BPNT 00123 034456 WIND: 00124'175400 INC 3.3 00125'000762 JMP RAIN

00126 102400	FINAL:	SUB	0.0		
00127 126400		SUB	1+1		
00130'006000-		JSR	e.PLOT	JRESET BEAM/PEN	TO LOWER
00131 .000000		Ø		I LEFT-HAND CORM	
00132'0060165		JSR	8.ALPH		
00133'032447		JMP	esv3	JEXIT	
00134'000000	FFR:	0			
00135'000207'	FFS:	FF			
00136'000225'	FFP:	LETT			
00137 '000001	NBLK:	1			
00140.000440	CORE:	440			
00141 000555	BLK:	555			
	1				
00142'054435	SING:	STA	3,583	JROUTINE TO PLOT	A BLOCK
00143'021001		LDA	0,1,2		
00144'025003		LDA	1,3,2		
00145'006000-		JSR	e.PLOT		
00146'177777		-1			
00147 021000		LDA	0,0,2		
00150'0240075		LDA	1. MSKR		
00151'107400		AND	0.1	INUMBER OF POINT	re internet and
00152'044426		STA	1,NPNTS	SHOHBER OF FOIN	1 Internet and the second
00153'126400					
		SUB	111		
00154'044427		STA	1.NP	ACET Y Y FOR FIL	ST DOINT
00155'0060035		JSR	e-PON1	JGET X,Y FOR FIR	
00156'040426		STA	0,X0	JREMEMBER THEM F	OR
00157 044426		STA	1,YØ	; LAST LINE.	
00160'006000-		JSR	e.PLOT	JPLOT A POINT	A STATE OF THE PARTY OF THE PAR
00161'000000		0		JBEAM OFF/PEN UP	
00162'000404		JMP	HAIL		NY INCOMENTATION IN CONTRACTOR IN
00163 0060045	FOG:	JSR	e.PON2	J2ND, QUICK ENTR	(1
00164'006000-		JSR	e.PLOT		
00165'000001		1		JBEAM ON / PEN D	NWD
00166'010415	HAIL:	ISZ	NP		
00167 024414		LDA	1,NP		
00170'014410		DSZ	NPNTS		
00171 000772		JMP	FOG		LAST POINT YET
00172 020412		LDA	0,X0	JGET FIRST POINT	BACK
00173 024412		LDA	1,40		
00174'006000-		JSR	e.PLOT	;PLOT IT	
00175'000001		1			
00176'002401		JMP	eSB3	JEXIT	
	3				
00177'000000	SB3:	Ø			
00200.000000	NPNTS:	Ø			
00201 '000030	BPNT:	Ø			
00202.000000	SV3:	Ø			
00203 . 000000	NP:	Ø			
00204'000000	XØ:	Ø			
00205.000000	YØ:	Ø			
00206'000000	CSV3:	Ø			
	TO PRI	NT "F" OI	N FIXED F	TITCKS	

JTO PRINT "F" ON FIXED BLOCKS 00207 054777 STA FF: 3,CSV3 LDA 0,1,2 00210.051001 1,3,2 LDA 00211'025003 3,FIVE LDA 00212'034411 00213'163000 ADD 3,0 00214'167000 3,1 ADD JEET BEAM POSITIONED 00215'006000e.PLOT JSR 00216.000000 Ø Montes

00217 0060165		JSR	e.ALPH	;ALPHA		
00220'0060065		JSR	e.PRN1	JPRINT	"F"	
00221 000106		"F				
00222'002764		JMP	ecsv3			
00223'000005	FIVE:	5	The state of the s			
00224'014000	FMSK:	14000				
00004 014000			ER ON PAL	PER		
00225'054432	LETT:	STA	3, SNOT			
00226'050433		STA	2,5V2			
00227 '030433		LDA	2, POINT			
00230'102400		SUB	0,0			
00231 '040417		STA	Ø,MODE			
00232'021000	PLOOP:	LDA	0,0,2	; (X:Y)		
00232 021000	FLOOF.	MOVS	0,1, SNR	,		
00234'000421		JMP	END			
00235 0340075		LDA	3. MSKR	- 14		
00236'167400		AND	3,1	JY		
00237'163400		AND	3.0	1X		
00240'151400		INC	2,2			
00241 050417		STA	2,112			
00242 030417		LDA	2,5V2			
00243'035001		LDA	3,1,2	JXG		
00244'163000		ADD	3,0	JXP		
00245'035003		LDA	3,3,2	JYG		
00246'167000		ADD	3,1	JYP		
00247 006000-		JSR	e.PLOT			
00250'000000	MODE:	Ø				
00251'102520		SUBEL	0.0			
00252'040776		STA	Ø,MODE			
00253'030405		LDA	2,172			
00254'000756		JMP	PLOOP			
00255'030404	END:	LDA	2,5V2			
00256'002401		JMP	eSNOT			
00257 000000	SNOT:	Ø				
00260'000000	IT2:	0				
00261 000000	SV2:	0				
00262'000263'	POINT:	.+1				
00263'007012		7012	ILETTER	"5"		
00264'007005		7005				
00265'002405		2405				
00266'005005		5005				
00267 005010		5010				
00270'000000		0				
00210 000000	; TO P	LOT LOAD	VECTORS			
00271 '0200015	LPLS:	LDA	Ø. PLTS			
00272 040000-	LILJ.	STA	Ø. PLOT			
00273 054572	DIRECT	STA	3,RVEC			
	DIREC:	A Transferration	All a strange of the second			
00274'0340055		LDA	3. MI			
00275'0200105		LDA	Ø. NUM			
00276'040563		STA	Ø,KNT			
00277 054563		STA	3.PNT			

00300'031400 REPT: 00301'021014 00302'101005 00303'000463 00304'021001 00305'025003 00306'006000-00307'000000 00310'025014 LDA MOV JMP LDA LDA JSR Ø LDA 2,0,3 0,14,2 0,0,SNR TRIP ;SKIP ERASED BLOCK 0,1,2 ;XC 1,3,2 ;YC 0.PLOT 1,14,2 ;WEIGHT

00311 044562		STA	1 . NW	C-117
00312:050551		STA	2,AC2	
00313'0060115		JSR	e.SCAL	
00314'030547		LDA	2,AC2	
00315'021001		LDA	0,1,2	; XC
00316'035003		LUA	3,3,2	JYC
00317'136400		SUB	1,3	and the second sec
00320'165000		MOV	3,1	
00321'006000-		JSR	.PLOT	
00322'000001		1		
00323'0060165		JSR	e.ALPH	
00324'020547		LDA	Øshh	
00325'0060145		JSR	8.IPRN	
00326'000004		4	e • 1 · / · · ·	
00327'030534		LDA	2,402	
00330'021001		LDA	0,1,2	JCENTROID AGAIN
00331'025003		LDA	1,3,2	JCENTROID AGAIN
00332'006000-		JSR	e.PLOT	
00333'000000		0	e . FLUI	
00334'025023		-	1 02 0	
		LDA	1,23,2	JX LOAD
00335'044536		STA	1,000	ICON F IT
00336'0060115		JSR	e.SCAL	ISCALE IT
00337'030524		LDA	2,402	
00340'021001		LDA	0,1,2	1 XC
00341'107000		ADD	0,1	
00342'044522		STA	1.XVEC	
00343'025024		LDA	1,24,2	JY LOAD
00344'044530		STA	1,VV	
00345'0060115		JSR	e.SCAL	
00346'030515		LDA	2,AC2	
00347'021003		LDA	0,3,2	JYC
00350'107000		ADD	0,1	
00351'020513		LDA	Ø, XVEC	JVECTOR NOW IN ACOJACI
00352'006000-		JSR	e.PLOT	
00353'000001		1		
and a second	;			A NUMBER OF THE FLAG TO PLOT MOT D
00354'020005-		LDA		J.NVEC IS THE FLAG TO PLOT/NOT B
00355'101005		MOV		ITHE MAG. OF APPLIED LOADS
00356'000410		JMP	TRIP	10 MEANS NO PLOT
	1			
00357'0060165		JSR	e.ALPH	
00360'020513		LDA	Ø,WW	
00361'0060145		JSR	e.IPRN	
00362'000004		4		
00363'020511		LDA	Ø, VV	
00364 0060145		JSR	e.IPRN	
00365'000004		4		
00366'010474	TRIP:	ISZ	PNT	
00367 034473		LDA	3, PNT	
00370'014471		DSZ	KNT	
00371 000707		JMP	REPT	
	3			

The second second preservices

JTO PRINT JOINT PRESSURES

00372'0300215 00373'151112 PLUM: 00374'002471 00375'025000 00376'0200075 00377'050467

3

LDA 2,.PRES MOVL# 2,2,SZC JMP @RVEC JEXIT LDA 1,0,2 JCONTROL WORD LDA 0,.MSKR STA 2,PR2

			100 100		
00400'123400		AND	1,0	INB	C-118
00401'106700		SUBS	0,1	INP	
00402'044465		STA	1,NPREM		
00403'0340055		LDA	3. • M1		
00404'117000		ADD	0.3		TAT 25 GAA (RAS)
00405'031400		LDA	2,0,3	JELOCK PO	INTER
00406'0060135		JSR	0.LENG		
00407'040451		STA	Ø,LENG		
00410'021014		LDA	0,14,2		
00411'101005		MOV	0,0,SNR		
00412'000442		JMP	FRED	SKIP ERAS	SED BLOCK
00413'0060035		JSR	e.PON1		
00414'040454		STA	Ø, XAA		
00415'044454		STA	1,YAA		
00416'024451		LDA	1.NPREM		
00417'125400		INC	1,1		
00420'021000		LDA	0,0,2	JCONTROL N	ND .
00421 0340075		LDA	3. MSKR		
00422'163400		AND	3,0	INC	
00423'106415		SUB#	2003	JCHECK FOR	R LAST CORNER
00424'126400		SUB	1,1		
00425'0060045		JSR	e.PON2		
00426 034442		LDA	3, XAA		
00427 163220		ADDER	3,0	3 (XA+XB) /2	2
00430'034441		LDA	3.YAA		Laboration Contractor
00431'167220		ADDER	3,1	1 (YA+YB)/2	2
00432'034440		LDA	3.NN5		
00433'162400		SUB	3,0		
00434'166400		SUB	3,1		
00435'0060015		JSR	e.PLTS		
00436'000000		Ø	e		
00437 0060165		JSR	e.ALPH		
00440'0060065		JSR	e.PRN1		
00440 00000052		"*	e .r Mai		
00441 000052		LDA	2, PR2		
00442 030424		LDA		FORCE	
			1,1,2	FUNCE	
00444'102440		SUBO	0,0		
00445'030412		LDA	2,N125		
00446'073301		MUL	0 1 540		
00447'030411		LDA	2, LENG		
00450'073101		DIV			
00451 121000		MOV	1,0		
00452'0060145		JSR	e.IPRN		
00453 000005		5			
00454 030412	FRED:	LDA	2, PR2		
00455'031002		LDA	2,2,2	JUINK	
00450'000715		JMP	PLUM		
000012		.RDX	10		
00457 000175	N125:	125	1. 1. 1. 1. 1. 1. 1. 1. 1. 1. 1. 1. 1. 1		
000010		•RDX	8		
00460 000000	LENG:	Ø			
00461 000000	KNT:	0			
00462.000000	PNT:	Ø			
00463'000000	AC2:	0			
001410000000	YUEC.	0			

00464'000000 XVEC: 00465'000000 RVEC: 00466'000000 PR2: 00467'000000 PR2: 00470'000000 NPREM: 00470'000000 XAA: 00471'000000 YAA:

00472'000005 NN5: 5 C-119 00473'000000 WW: Ø 00474'0000000 VV: 0 *END

		.TITL	CONTR	
	; DYNAMI	C ITERAT	ION CONT	ROL ROUTINE
		.ENT	CONTRA	PFLG, C100, VEC, LPAP, UREP
		.EXTD	.OVL .GI	ETT, DISS, MOT, CURS, PRNI, HITC
		.EXTD	.PLTS	PAGE, ALLB, FORD, MI, NUM, CPNI
		.EXTD	.DISP	SCAL, . LPLS, . VFAC, MU, . RLNC, . UINP
		.EXTD	.REBE	EMPT PON1 PON2 MSKR M3 M5
		.EXTD	. INPH.	ITS, .PEN2, .ALPH, .TYP, .LENG, .MESS
		.EXTD	.PSEG	DISB, . IPRN, . READ, . WRIT, . STEP, . TPN
		.EXTD	.LODE	DCM. MOVE RSET KET TIME
		.EXTN	OPTIN	
		. ZREL		
00000-000000	.LPAP:	Ø	JHARD CO	OPY LOAD-PLOT FLAG
00001-000000	.VEC:	Ø	; VECTOR	PLOT FLAG (1=PLOT, 0=DUN'T)
00002-000000	.PFLG:	Ø		
00003-000100	.C100:	100		
00004-000023	.UREP:	23	UPDATE	FREQUENCY
00004 000000	· on Li	.NREL		
00000'000000	UCNT:	0		
00000 000000	:			
	:MA	IN CALCI	LATION C'	CL F
	:			To be a second
00001 1020004-	GRUNT:	LDA	Ø. UREP	
00002'040776	onon	STA	Ø,UCNT	
00003'0060045	DYN:	JSR	e.MOT	JLAW OF MOTION
00004'0060575		JSR	P.KET	JK.E.ROUTINE
00005 0060135		JSR	e.FORD	FORCE/DISPLACEMENT LAW
00006'0060515		JSR	e.STEP	J INCREMENT CYCLE COUNTER
00007 '0060545		JSR	e.DCM	JDISP MACHINE
00010'063710		SKPDZ	TTI	
00011'004407		JSR	OUT	JKEY HAS BEEN HIT
00012'014766		DSE	UCNT	
00013'000770		JMP	DYN	
00014'0060125		8		JUPDATE CONTACT LIST
00015'000764	No. of Concession, Name	JMP	GRUNT	VOLDATE CONTROL ELOT
00010 000104		0.11	CROIT	
	1			
	1			
00016'000257'	RT3:	RET3		
00017'100257'		eRET3		
00020'056776		STA	3, eRT3	
00021 0060405		JSR		
00022'060510		DIAS		JGET KEY CHARACTER
00023'030426		LDA		POINTER TO KEY LIST
00024 000403		JMP	SEEK	
00025'151400	NEXT:	INC	2,2	
00026'151400		INC	2,2	
00027 025000	SEEK:	LDA	1,0,2	
00030'125015		MOV#		CHECK FOR LIST END
00031'002766		JMP	eRTT3	CHARACTER NOT FOUND
00032 034413		LDA	3.MSK	RIGHT 7 RITS

00032'034413 00033'163400 00034'137400 00035'162414 00036'000767 00037'166405 00040'003001 00041'034407 00042'166415 00043'003001 LDA AND SUB# JMP SUB JMP LDA SUB# JMP 3,MSK ;RIGHT 7 BITS 3,0 1,3 ;JUST CHARACTER ALONE 3,0,SER NEXT ;NOT THIS ONE 3,1,SNR ;FOUND IT! GET FLAG IN AC1 01,2 ;GO TO APPROPRIATE ROUTINE 3,STATU ;STATUS FLAG 3,1,SNR ;IS PERMISSION GRANTED? 01,2 ;YES. GO TO ROUTINE

-										
	00044'002753		JMP	ORTT	3	BACK H	ROM	HENCE	YOU C	AME
	00045'000177	MSK:	177							
	00046'100000	RFLAG:	100000							
	00047 . 640000	SFLAG:	40000							
	00050'000000	STATU:	Ø							
	00051 '000052'	POINT:	•+1							
		3								
		ILIST OF	F POSSIBI	LE KE	YS 1	THAT CAN	N BE	HIT		
		;								
	00052.000104		"D							
	00053'000166'		USELY		DRAI	" BLUCK	5			
	00054'040120		"P+4000		TO 1					
	00055'000135'		PHASE	the first state and the	10 1	PHASE I				
	00057 000132		"G+40000 GO		OT 1	NANTOS	-			
	00060'100123		"S+10000		RI 1	JINAMICS	5			
	02061 '000124'		STOP	and the second se	P D	NAMICS				
	00062'000132		"2	,510	1 0	11141105				
	00063'000172'		ZERO	ISET	AL 1	VELOCI	TIES	TO FE	2.11	
	00064'100116		"N+10000		1					
	00065 000156		NOPLT		SE S	SCREEN &	SUP	PRESS I	PLOTTI	ING
	00066'100101		"A+1000		1778-1778-198			A 19 22 47.450 A		1996 4 97
	00067'000162'		ACTIV	JACT	IVAT	TE PLOTT	TING	AGAIN		
	00070'040111		"I+40000	3						
	00071'000210'		INPUT	JINP	UTI	ATA				
	00072'000110		"н							
	00073'000252'		HARD	JMAK	E HA	ARD COPY	(
	00074'000126		···v	20.000.000			87			
	00075'000260'		VEC	; VEC	TOR	DISPLAY	(
	00076'000114		"L	1.0.						
	00077'000271'		LPLOT	310	PLOI	LOADS	ONLY			
	00100'000124		"T		-	-	TVD			
	00101'000275'		TYPEN		PRI	VI PRUP	. 11P	E # D		
	00102'040112		"J+40000 PINP		TNP	T IOINT	PPF	SSURE		
	00103'000417' 00104'040122		"R+40000		TIVEL	51 501141	TAL	SSORE		
	00105'000425'		RP3		REAL	A P-3	FILE			
	00106'040127		"W+40000							
	00107'000432'		WP3		WRIT	E A P-S	3 FIL	E		
	00110'040103		"C+40000							
	00111'000434'		CUR		UP	CURSOR	AND	HAIT		
	00112'040130		"X+40000							
	00113'000151'		RESET	; TO	RESE	T CYCLE	cou	NTERS, B	TC	
	00114'040121		"0+4000							
	00115'000150'		TIME		CHAN	GE DYN	FACS			
	00116'040115		"M+40000							
	00117'000145'		MOVM		SET	DISP CC	DNTRO	L		
	00120'040102		"B+40000				-	OCUS		
	00121'000146'		BOLT				E BL	UCKS		
	00155.000000		Ø	JEND	OF	LIST				
	001001000401	;	IMD	STOP						
	00123'000401	CONTR:	JMP	STOP						
	00124'020723	STOP:	LDA	Ø,SF	LAG					
	00125'040723		STA	and the state of the		"STOP	' STA	TUS		
	00126'063610		SKPDN	TTI		SWAIT F				
	00127'000777		JMP	1						
	00130'004670		JSR	OUT						
	00131 000773		JMP	STOP	8					
		;								

 00132 020714	GQ:	LDA	Ø, RELAG	
00133'040715	00.	STA	and the second s	"RUN" STATUS
00134'000645		JMP	GRUNT	, how office
00134 000043		5.11	UNUNI	
001051040477	PHASE:	READS	Ø	CANT LEAVE W/0-UP
00135'060477	PHASE:		0,0,520	CANT LEAVE WID-DI
00136'101122		MOVEL	With the second second	
00137'000765		JMP	STOP	
00140 0060115		JSR	e.PAGE	
00141 102520		SUBEL	0.0	Contract on the
00142'0060015		JSR	e.ovl	JOVERLAY #1
00143'063077		HALT		;TAPE ERROR
00144'000775		JMP	3	
	;			
00145'0020555	MOVM:	JMP	e-MOVE	
	;		-	
00146'063077	BOLT:	HALT		
00147'000755		JMP	STOP	
	;			
00150'0060605	TIME:	JSR	0.TIME	
	;			
00151 0060565	RESET:	JSR	0.RSET	
00152'0060115		JSR	0.PAGE	
00153'0060525		JSR	e.TPRN	
00154'0060035		JSR	e.DISS	
00155'002502		JMP	eRET3	
	3			
00156'0060115	NOPLT:	JSR	0.PAGE	
00157'102520		SUBEL	0,0	
00160 040002-		STA	Ø. PFLG	SUPPRESS PLOTTING
00161'002476		JMP	eRET3	
	;			
00162'102400	ACTIV:	SUB	0.0	
00163'040002-		STA		RE-ACTIVATE PLOTTING
00164'0060525		JSR	0.TPRN	WRITE NO. OF ITERATIONS
00165'002472		JMP	PRET3	A.A
	;		OCC/IP.	
00166'0060115	DSPLY:	JSR	e.PAGE	FRASE SCREEN
00167 0060525		JSR	0.TPRN	WRITE NO. OF ITERATIONS
00170'0060035		JSR	e.DISS	FRE-DRAW SYSTEM
00171'002466		JMP	ORET3	
00111 002400	1			
00172 0300145	ZERO:	LDA	2 M1	
00173'0240155	ELING.	LDA	1 NUM	
00174'124400		NEG	1.1	
00175'102400		SUB	0.0	
00176 035000	ITER:	LDA	3,0,2	
00177 041405		STA	0,5,3	X-VEL
00200'041406		STA	0,6,3	FALPHA-DOT
00201 041408		STA	0,15,3	FY-VEL
00202'151400		INC	2,2	and the second second
00202 151400		1110	1 1 000	

00203'125404 00204'000772 00205'0060065 00206'000007 00207'002450 INC 1,1,SER JMP ITER JSR 0.PRN1 7 #RING BELL JMP 0RET3

; INPUT ROUTINE -- FRICTION, LOADS, UNITS & OPTIONS

00210'0060435 INPUT: JSR 0.MESS 00211'001617' INMS

3

2					C-123
	00212'177324		-300.		and the second
	00213'001212	DOUDD.	650 .	-	THURS OCL
	00214'0060025	DOVER:	JSR		JWAIT FOR CHAR
	00215 024426		LDA	1. CRGRT	
	00216'106415		SUB#	Ø, I, SNR	C. III MACONA
	00217 002440		JMP	efet3	CHANGED YOUR MIND
	002201024424		LDA	1. CHEF	
	07221'1 0.14		SUB#	ColozeR	
	00222'000403		JMP	•+3	A LN PSD * CDCNW
	00223'0060355		JSR		J GO TO INPUT FRICTION
	00224'002433		JMP	eRET3	
	00225'024420		LDA	1, CHRU	
	00226'106414		SUB#	Ø,1,52R	
	00227 000403		JMP	•+3	
	00230'0060255		JSR		JGO TO INPUT UNITS
	00231 1002426		JHF	eret3	
	00232 024414		LDA	1.ChnL	
	00233'106414		SUB#	0,1,SER	
	66234'000403		JMP	•+3	
	00235'086414		JSR	elodo	3 GO TO INPUT LOADS
	00236'002421		JMP	ØRET3	
	00237 024410		LDA	1.CHRO	
	00240'106415		SUB#	0,1,5NR	
	00241 002407		JMP	COPTNN	; GO TO SET OPTIONS
	00242'000752		JMP	DOVER	; DO IT OVER
	00243'000015	CRGRT:	15		
	00244'000106	CHRF:	"F		
	00245'000125	CHRU:	"U		
	00246'000114	CHRL:	"L		
	00247'000117	CHRO:	"0		
	00250'177777	OPTNN:	OPTIN		
	00251'001121'	LODO:	ONLY		
		;			States and a state of the second
		;HARD:	READS	Ø	CHECK FOR SW. 0
		;	MOVEL		;OFF=4631,ON=PLOTTER
		;	JMP	PLTR	
	00252'0060065	HARD:	JSR	e.PRN1	10
	00253'000033		27.		JASCII ESC
	00254'0060065		JSR	e . PRN1	ALASSO TOURS
	00255'000027		23.		JASCII ETB
	00256'002401		JMP	eRET3	
		;PLTR:	JSR	e.DISP	
		;	JMP	eret3	
		;			
	00257 000000	RET3:	0		
		3			
	00260'102520	VEC:	SUBEL	0.0	M. ISTRACTOR
	00261'040001-		STA		SET VECTOR PLOT FLAG
	00262*0060045		JSR	e.MOT	
	00263'0060575		JSR	e.KET	
	00264'0060135		JSR	e.FORD	JONE SCAN FOR PLOTTING
			1 C 2 C 2 C	Contraction of the second seco	

JSR 00265 0060515 SUB 00266 102400 STA 00267 * 040001-JMP 00270'002767 ---; --00271 '0060215 LPLOT: JSR SUBEL 00272'102520 STA 00273 '040000-JMP 00274'002763

@.STEP #INCREMENT CYCLE COUNTER
0,0
0,.VEC #KNOCK DOWN FLAG
@RET3 #EXIT

C 100

@.LPLS #TO PLOT LOADS 0,0 #SET HARD COFY FLAG 0,.LPAP @RET3

*					6-124	
		;				
		;TO PRIM	NT TYPE	#'S ON BLOCK	EDGES	
	00275 0340145	TYPEN:	LDA	3		
	00276'054502		STA	3,BLOCK		
		SCAN BI	LOCKS	-		
	00277 031400	BEGIN:	LDA	2,0,3		
	00300 151005		MOV	2,2, SNR		
	00301 002756		JMP	eRET3		
	00302'021014		LDA	8,14,2		
	00303'101005		MOV	0.0.SNR		
	00304'000440		JMP	NEXT1		
	000004 000440	SCAN S	IDES	I CANTA		
	00305'021000	JEGAN S.	LDA	0,0,2		
			LDA	1. MSKR		
	00306'0240325					
	00307'107400		AND	Ø . I		
	00310'044471		STA	1,NPNTS		
	00311'126400		SUB	1,1		
	00312'044470		STA	1,NPP		
	00313'0060305		JSR	e.PON1		
	00314'040467		STA	Ø, XØ		
	00315'040470		STA	0, XA		
	00316'044466		STA	1,40		
	00317 044470		STA	1,YA		
	00320'024462		LDA	1,NPP		
	00321 000414		JMP	DOWN		
	00322'125400	BACK:	INC	1.1		
	00323'0060315		JSR	e.PON2		
	00324'040462		STA	Ø,X8		
	00325'044463		STA	1,YB		
	00326'004421		JSR	TPRNT		
	00327 010453		ISZ	NPP		
	00330'024452		LDA	1,NPP		
	00331 020455		LDA	Ø,XB		
	00332'040453		STA	Ø,XA		
	00333'020455		LDA	Ø,YB		
	00334'040453		STA	Ø,YA		
	00335'014444	DOWN:	DSE	NPNTS		
	00336'000764		JMP	BACK		
	00337 020444		LDA	Ø, XØ		
	00340'040446		STA	0,XB		
	00341 020443		LDA	0,Y0		
	00342'040446		STA	Ø,YB		
	00343'004404		JSR	TPRNT		
		;END OF	SIDE S	CAN		
	00344'010434	NEXT1:	ISE	BLOCK		
	00345'034433		LDA	3,BLOCK		
	00346'000731		JMP	BEGIN		
		JEND OF	BLOCK	SCAN		
		3				
	00347 054430	TPRNT:	STA	3, TPSAV		
	00350'024432		LDA	1,NPP		
	0005110010.10		100	0 TUD	F TURE # TURE PROF	

00351'0060415 00352'101005 00353'002424 00354'040435 00355'020430 00356'034430 00356'034430 00357'163220 00360'034432 00361'162400 JSR MOV JMP STA LDA LDA ADD2R LDA SUB @.TYP ;GET TYPE #, THIS EDGE Ø.Ø.SNR ;DEFAULT @TPSAV Ø.TYPE Ø.XA 3.XB 3.0 ;(XA+XB)/2 3.MOVE1 3.0

					U-1
00362 024425		LDA	1,YA		
00363'034425		LDA	3,YB		
00364'167220		ADDER	3,1	; (YA+YB)/2	
003651034425		LDA	3. MOVEL		
00366'166400		SUB	3 . 1		
367 0060105		Jon	C.PLTS		
00378'000000		0			
00371 '0060405		JSR	P.ALPH		
00372 020417		LDA	O, TYPE		
00373'034420		LDA	3,NN0		
00374'163000		ADD	3,0	JASCII CHAR	
00375'0060375		JSR	8. FRN2		
00376'002401		JMP	OTFSAV		
00377'000009	TPSAV:	Ø			
09430'000000	BLOCK:	0			
01401 1000000	NPNTS:	0			
004021000000	Miler :	1			
Re403*P00000	XC:	Ø			
00404'000000	Y0:	0			
00405'000000	XA:	Ø			
00436'3300000	XB:	0			
00407.000000	YA:	0			
00410'000000	YB:	Ø			
00411 '000000	TYPE:	Ø			
00412'000006	MOVE1:	6			
00413 000060	NNØ:	"Ø			
00414'001100'	FLG:	FLAG			
	;				
00415 0060255	UINP:	JSR	0.UINP		
00416'002641		JMP	ØRET3		
	3				
00417 0060435	PINP:	JSR	e.MESS		
000012		.RDX	10		
00420'001461'		PMESS			
00421 177324		-300			
00422'001274		700			
000010		.RDX	8		
00423'0060445		JSR	8.PSEG		
00424'002633		JMP	eRET3		
	;				
00425'0060475	RP3:	JSR	e.READ		
00426'0060115		JSR	0.PAGE		
00427 0060525		JSR	0.TPRN		
00430 0060035		JSR	e.DISS		
00431 '002626		JMP	eRET3		
	3				
00432'0060505	WP3:	JSR	8.WRIT		
00433'002624		JMP	eRET3		
and a second second	;				
00434'102400	CUR:	SUB	0.0		
221051010257		STA	0.0516	:RESET PROP.	CHNG. IN

e.CURS

Ø, OFLG ; RESET PROP. CHNG. INDIC.

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00435'042757 00436'0060055 CURS: 00437 000522' 00440 000641 00441'000642' 00442'0060405 00443'020457 00444 024462 00445'106415 00446 002456

JSR CHAR X Y JSR LDA LDA SUB# JMP

STA

P.ALPH Ø, CHAR 1,01 C.1. SNR :"1" BEEN HIT? OLOADR

-					
	00447 024464		LDA	1,0	
	00450'106415		SUB#	D.1.SNR	; HAS "O" BEEN HIT ?
	00451 002454		JMP	CONE	
	00452'024456		LDA	1,0	
	00453'106415		SUB#	Ø,1,SNR	; HAS "U" BEEN HIT?
	00454'000575		JMP	UNFIX	;YES
	00455'024455		LDA	1,E	
	00456'106415		SUB#	0,1,SNR	; HAS "E" BEEN HIT?
	00457 000455		JMP	ERASE	;YES
	00460'024451		LDA	1.F	
	00461'106414		SUB#	Ø,1,SER	; HAS "F" BEEN HIT?
	00462'002441		JMP	e SURFR	JTRY PROPERTY KEYS
	00463'0060075		JSR	e.HITC	
	00464'002641'		X		
	00465'000642'		Y		
	00466 900750		JMP	CURS	
	00467 021000		LDA	0,0,2	CONTROL WORD
	00470 024427		LDA	1,FBIT	;"FIXED" FLAG (BIT 3)
	00471 107414		AND#	0,1,SER	JALREADY FIXED?
	00472'000744		JMP	CURS	
	00473'123000		ADD	1,0	JADD IN FLAG
	00474'041000		STA	5,9,0	PUT WORD BACK
	00475'102400		SUB	0.0	SUPPRESS VELOCITIES
	00476'041005		STA	0,5,2	X-VEL
	00477 041006		STA	0,6,2	JALPHA-DOT
	00500 041015		STA	0,15,2	;Y-VEL
	00501 '041020		STA	0,20,2	;DELTA-X
	00502'041021		STA	0,21,2	;DELTA-Y
	00503'041022		STA	0,22,2	;DELTA-ALPHA
	00504'034415		LDA	3,FIVE	
	00505'021001		LDA	0,1,2	JXC
	00506'163000		ADD	3,0	\$XC+5
	00507 025003		LDA	1,3,2	JYC
	00510'167000		ADD	3,1	;YC+5
	00511'0060105		JSR	e.PLTS	BC ADDING ATTING
	00512.000000		Ø		PUT BEAM TO RIGHT PLACE
	00513'0060405		JSR	e.ALPH	
	00514'0060065		JSR	e.PRN1	
	00515'000106		"F	And the second second	
	00516'000720		JMP	CURS	
	00517 010000	FBIT:	10000	The second s	FIX BIT
	00520'004000	MBIT:	4000		FIX BIT
	00521 000005	FIVE:	5		
	00522 000000	CHAR:	Ø		
	00523'001020'	SURFR:	SURF		
	00524'000672'	LOADR:	LOAD		
	00525'001121'	ONE:	ONLY		
	00526'000261	C1:	"1+200		
	00527 000262	C2:	"2+200		
	COULT CONLOC				

COVER CONCORE				
00530'000325	U:	"U+200		
00531 '000306	F:	"F+200		
00532'000305	E:	"E+200		
00533'000317	0:	"0+200		
00534'0060075	ERASE:	JSR	e.HITC	
00535'000641'		X		
00536'000642'		Y		
00537 000677		JMP	CURS	INO
00540'044503		STA	1.NB	;BL
00541 0060115		JSR	e.PAGE	
00542'0060265		JSR	e.REBE	; PU

D HIT LOCK # DT IN CORRECT BOXES

0		1	-
	- 1	1	1
		-	1

00543'102400		SUB	0.0		
00544 041014		STA	0,14,2	SET AREA	IO ZERO
00545 021000		LDA	0,0,2		
00546'0240325		LDA	1. MSKR		
00547'123400		AND	1,0		
00550'040477		STA	0, PCNT		
00551 126400		SUB	1,1		
00552'044472		STA	1,NP		
	INEXT P	ART REMON	VES ALL I	POINT ENTRIE	D F. WH
	BOX AR				
00553'0060305		JSR	e.PON1		
00554'000403		JMP	PLACE		
00555'024467	COW:	LDA	1.NP		
00556'0060315		JSR	e.PON2		
00557 0340335	PLACE:	LDA	3 . · M3		
00560'030003-		LDA	20100		
00561 040465		STA	ØINX		
00562'102400		SUB	0.0		
00563'073101		DIV			
00564'127120		ADDZL	1,1		
1 6565'12712		ADDEL	1 . 1		
00566'137030		ADD	1,3		
00567 024457		LDA	LINX		
00570'102400		SUB	0,0		
00571'073101		DIV			
00572'137000		ADD	1,3		
00573'054452		STA	3,OLD		
00574'020447		LDA	Ø,NB		
00575 024447		LDA	1.NP		
00576 125300		MOVS	1 . 1		
00577'123000		ADD	1,0	; (NP:NB)	
00600 035400		LDA	3,0,3	SCNO CHECK	FOR END)
00601 025400	ROUND:	LDA	1,0,3		
00602'106415		SUB#	0,1, SNR		
00603 000405		JMP	OOT	FOUND IT	
00604'165400		INC	3,1		
00605'044440		STA	1,OLD		
00606 035401		LDA	3,1,3	SLINK	
00607 000772		JMP	ROUND		
00610'025401	00T:	LDA	1,1,3	;THIS LINK	
00611 046434		STA	1,00LD		
00612'010432		ISZ	NP		
00613'014434		DSZ	PCNT		
00614'000741		JMP	COW		
	TO RETU	JRN DEAD	CONTACT	ENTRIES TO	EMPTY LIST
00615 0340345		LDA	3		
00616'020425		LDA	Ø,NB		
00617'117000		ADD	0,3		
00620 054425		STA	3,0LD		
00621 035400		LDA	3,0,3		
antool L/FCOO		MON	2.1	INFER FIRST	FNTRY

00622'165000 00623'175112 00624'000411 00625'171000 NIT: 00626'035402 00627'175113 00630'000775 00631'056414 00632'0200275 00633'041002

3,1 JKEEP FIRST ENTRY MOV MOVL# 3,3,SEC INO CONTETS JMP EXIT 3,2 ;SAVE PREV. ADDR. (LAST?) MOV INEXT ENTRY 3,2,3 LDA 3,3, SNC MOVL# NIT SKEEP GOING DOWN CHAIN JMP 3,00LD ; PLUG INITIAL POINTER STA O. EMPT LDA 0,2,2 ;STORE OLD EMPT POINTER STA

00634'0440275		STA	1. EMPT	AUDDATE DEMAINING CONTACTS
00635'0060125	EXII:	JSR	e.ALLB	JUPDATE REMAINING CONTACTS
00636 0060525		JSR	e.TPRN	ADC DOAL
00637 0060035		JSR		;RE-DRAW
00640'002410		JMP	ecursr	
00641 000000	X:	0		
00642.000000	Y:	0		
00643'000000	NB:	Ø		
00644'000000	NP:	Ø		
00645'000000	OLD:	Ø		
00646'000000	NX:	Ø		
00647 .000000	PCNT:	Ø		
00650'000436'	CURSR:	CURS		
00651 '0060075	UNFIX:	JSR	e.HITC	
00652'000641'		X		
00653'000642'		Y		
20654 02774		JMP	ecursr	
00655'021000		LDA	0,0,2	TO RELEASE A BLOCK
00656'024642		LDA	1,MBIT	; IS MASTER BIT SET?
00657 107414		AND#	0,1,52R	
00660'002770		JMP	OCURSR	;YES, HARD LUCK!
00661 024636		LDA	1.FBIT	
00662'107415		AND#	Ø,1,SNR	FIXED ALREADY?
00663'002765		JMP	CURSR	INO CHANGE NECESSARY
00664'122400		SUB	1,0	REMOVE BIT
00665 041000		STA	0,0,2	FUT CONTROL WORD BACK
I HE REAL PROPERTY AND A REAL PROPERTY AND A REAL PROPERTY.		JSR	e.PAGE	
10000 0000115				
00666'0060115		JSR	0.TPRN	
00667'0060525		JSR JSR	e.TPRN e.DISS	JRE-DRAW
	;	JSR JMP	e.DISS ecursr	JRE-DRAW JCARRY ON
00667'0060525 00670'0060035 00671'002757	;	JSR JMP E TO INPU	e.DISS ecursr JT LOAD	
00667'0060525 00670'0060035 00671'002757 00672'0060075	;	JSR JMP E TO INPU JSR	e.DISS ecursr	JCARRY ON
00667'0060525 00670'0060035 00671'002757 00672'0060075 00673'000641'	;	JSR JMP E TO INPU JSR X	e.DISS ecursr JT LOAD	JCARRY ON
00667'0060525 00670'0060035 00671'002757 00672'0060075 00673'000641' 00674'000642'	;	JSR JMP E TO INPU JSR X Y	e.DISS eCURSR JT LOAD V e.HITC	JCARRY ON VECTORS FROM SCREEN
00667'0060525 00670'0060035 00671'002757 00672'0060075 00673'000641' 00674'000642' 00675'000521	;	JSR JMP E TO INPU JSR X Y JMP	e.DISS eCURSR JT LOAD V e.HITC SURF1	JCARRY ON
00667'0060525 00670'0060035 00671'002757 00672'0060075 00673'000641' 00674'000642' 00675'000521 00676'050501	;	JSR JMP E TO INPO JSR X Y JMP STA	e.DISS eCURSR JT LOAD V e.HITC SURF1 2,PNT1	JCARRY ON VECTORS FROM SCREEN JNO HIT; TRY SURFACE
00667'0060525 00670'0060035 00671'002757 00672'0060075 00673'000641' 00674'000642' 00675'000521 00675'000521 00676'050501 00677'0060065	;	JSR JMP E TO INPU JSR X Y JMP STA JSR	e.DISS eCURSR JT LOAD V e.HITC SURF1	JCARRY ON VECTORS FROM SCREEN
00667'0060525 00670'0060035 00671'002757 00672'0060075 00673'000641' 00674'000642' 00675'000521 00675'000521 00677'0060065 00700'000007	;	JSR JMP E TO INPO JSR X Y JMP STA JSR 7	e.DISS eCURSR JT LOAD e.HITC SURF1 2.PNT1 e.PRN1	JCARRY ON VECTORS FROM SCREEN JNO HIT; TRY SURFACE
00667'0060525 00670'0060035 00671'002757 00672'0060075 00673'000641' 00674'000642' 00675'000521 00675'000521 00677'0060065 00700'000007 00701'0060055	;	JSR JMP E TO INPU JSR X Y JMP STA JSR 7 JSR	e.DISS eCURSR JT LOAD V e.HITC SURF1 2,PNT1	JCARRY ON VECTORS FROM SCREEN JNO HIT; TRY SURFACE
00667'006052S 00670'006003S 00671'002757 00672'006007S 00673'000641' 00674'000642' 00675'000521 00675'000521 00677'006006S 00700'000007 00701'006005S 00702'000522'	;	JSR JMP E TO INPU JSR X Y JMP STA JSR 7 JSR 7 JSR CHAR	e.DISS eCURSR JT LOAD e.HITC SURF1 2.PNT1 e.PRN1	JCARRY ON VECTORS FROM SCREEN JNO HIT; TRY SURFACE
00667'0060035 00670'0060035 00671'002757 00672'0060075 00673'000641' 00674'000642' 00675'000521 00675'000521 00677'0060065 00700'000007 00701'0060055 00702'000522' 00703'001000'	;	JSR JMP E TO INPU JSR X Y JMP STA JSR 7 JSR 7 JSR CHAR XX	e.DISS eCURSR JT LOAD e.HITC SURF1 2.PNT1 e.PRN1	JCARRY ON VECTORS FROM SCREEN JNO HIT; TRY SURFACE
00667'0060525 00670'0060035 00671'002757 00672'0060075 00673'000641' 00674'000642' 00675'000521 00675'000521 00677'0060065 00700'000007 00701'0060055 00702'000522' 00703'001000' 00704'001001'	;	JSR JMP E TO INPU JSR X JMP STA JSR 7 JSR 7 JSR CHAR XX YY	0.DISS 0CURSR JT LOAD 0.HITC SURF1 2.PNT1 0.PRN1 0.CURS	JCARRY ON VECTORS FROM SCREEN JNO HIT; TRY SURFACE
00667'0060525 00670'0060035 00671'002757 00672'0060075 00673'000641' 00674'000642' 00675'000521 00675'000521 00677'0060065 00700'000007 00701'0060055 00702'000522' 00703'001000' 00704'001001' 00705'0060405	;	JSR JMP E TO INPU JSR X Y JMP STA JSR 7 JSR 7 JSR CHAR XX YY JSR	e.DISS eCURSR JT LOAD V e.HITC SURF1 2.PNT1 e.PRN1 e.CURS	JCARRY ON VECTORS FROM SCREEN JNO HIT; TRY SURFACE
00667'0060525 00670'0060035 00671'002757 00672'0060075 00673'000641' 00674'000642' 00675'000521 00675'000521 00677'0060065 00700'000007 00701'0060055 00702'000522' 00703'001000' 00704'001001' 00704'001001'	;	JSR JMP E TO INPU JSR X Y JMP STA JSR 7 JSR 7 JSR CHAR XX YY JSR LDA	e.DISS eCURSR JT LOAD e.HITC SURF1 2.PNT1 e.PRN1 e.CURS e.ALPH Ø.CHAR	JCARRY ON VECTORS FROM SCREEN JNO HIT; TRY SURFACE
00667'0060525 00670'0060035 00671'002757 00672'0060075 00673'000641' 00674'000642' 00675'000521 00675'000521 00677'0060065 00700'000007 00701'0060055 00702'000522' 00703'001000' 00704'001001' 00705'0060405 00706'020614	;	JSR JMP E TO INPU JSR X Y JMP STA JSR 7 JSR CHAR XX YY JSR LDA LDA	e.DISS eCURSR JT LOAD V e.HITC SURF1 2,PNT1 e.PRN1 e.CURS e.CURS e.ALPH ø.CHAR 1,C2	JCARRY ON VECTORS FROM SCREEN JNO HIT; TRY SURFACE JRING BELL FOR HIT
00667'0060525 00670'0060035 00671'002757 00672'0060075 00673'000641' 00674'000642' 00675'000521 00675'000521 00677'0060065 00700'000007 00701'0060055 00702'000522' 00703'001000' 00704'001001' 00705'0060405 00706'020614 00707'024620 00710'106414	;	JSR JMP E TO INPU JSR X Y JMP STA JSR 7 JSR 7 JSR CHAR XX YY JSR LDA LDA SUB#	e.DISS eCURSR JT LOAD e.HITC SURF1 2.PNT1 e.PRN1 e.CURS e.ALPH Ø.CHAR 1.C2 Ø.1.SZR	JCARRY ON VECTORS FROM SCREEN JNO HIT; TRY SURFACE JRING BELL FOR HIT JIS IT "2" FOR 2ND POINT?
00667'0060525 00670'0060035 00671'002757 00672'0060075 00673'000641' 00674'000642' 00675'000521 00675'000521 00677'0060065 00700'000007 00701'0060065 00702'000522' 00703'001000' 00704'001001' 00705'0060405 00706'020614 00707'024620 00710'106414 00711'002737	;	JSR JMP E TO INPU JSR X Y JMP STA JSR 7 JSR CHAR XX YY JSR LDA LDA LDA SUB# JMP	e.DISS eCURSR JT LOAD V e.HITC SURF1 2,PNT1 e.PRN1 e.CURS e.CURS e.ALPH 0,CHAR 1,C2 0,1,SZR eCURSR	JCARRY ON VECTORS FROM SCREEN JNO HIT; TRY SURFACE JRING BELL FOR HIT
00667'0060525 00670'0060035 00671'002757 00672'0060075 00673'000641' 00674'000642' 00675'000521 00675'000521 00677'0060068 00700'000007 00701'0060065 00702'000522' 00703'001000' 00704'001001' 00705'0060405 00706'020614 00707'024620 00710'106414 00711'002737 00712'0060075	;	JSR JMP E TO INPU JSR X Y JMP STA JSR 7 JSR 7 JSR CHAR XX YY JSR LDA LDA SUB# JMP JSR	e.DISS eCURSR JT LOAD e.HITC SURF1 2.PNT1 e.PRN1 e.CURS e.ALPH Ø.CHAR 1.C2 Ø.1.SZR	JCARRY ON VECTORS FROM SCREEN JNO HIT; TRY SURFACE JRING BELL FOR HIT JIS IT "2" FOR 2ND POINT?
00667'0060525 00670'0060035 00671'002757 00672'0060075 00673'000641' 00674'000642' 00675'000521 00675'000521 00677'0060065 00700'000007 00701'00600055 00702'000522' 00703'001000' 00704'001001' 00705'0060405 00706'020614 00707'024620 00710'106414 00711'002737 00712'0060075 00713'001000'	;	JSR JMP E TO INPU JSR X Y JMP STA JSR 7 JSR CHAR XX YY JSR LDA LDA LDA SUB# JMP JSR XX	e.DISS eCURSR JT LOAD V e.HITC SURF1 2,PNT1 e.PRN1 e.CURS e.CURS e.ALPH 0,CHAR 1,C2 0,1,SZR eCURSR	JCARRY ON VECTORS FROM SCREEN JNO HIT; TRY SURFACE JRING BELL FOR HIT JIS IT "2" FOR 2ND POINT?
00667'0060525 00670'0060035 00671'002757 00672'0060075 00673'000641' 00674'000642' 00675'000521 00675'000521 00677'0060065 00700'0000007 00701'0060005 00702'000522' 00703'001000' 00704'001001' 00705'0060405 00706'020614 00707'024620 00710'106414 00711'002737 00712'0060075 00713'001000' 00714'001001'	;	JSR JMP E TO INPU JSR X Y JMP STA JSR 7 JSR CHAR XX YY JSR LDA LDA SUB# JMP JSR XX YY	e.DISS eCURSR JT LOAD V e.HITC SURF1 2,PNT1 e.PRN1 e.CURS e.CURS e.ALPH 0,CHAR 1,C2 0,1,SZR eCURSR e.HITC	JCARRY ON VECTORS FROM SCREEN JNO HIT; TRY SURFACE JRING BELL FOR HIT JIS IT "2" FOR 2ND POINT? JNO, SOMETHING ELSE
00667'0060525 00670'0060035 00671'002757 00672'0060075 00673'000641' 00674'000642' 00675'000521 00675'000521 00677'0060065 00700'000007 00701'0060055 00702'000522' 00703'001000' 00704'001001' 00705'0060405 00710'106414 00711'002737 00712'0060075 00713'001000' 00714'001001' 00715'000422	;	JSR JMP E TO INPU JSR X Y JMP STA JSR 7 JSR CHAR XX YY JSR LDA LDA LDA SUB# JMP JSR XX YY JMP	e.DISS eCURSR JT LOAD e.HITC SURF1 2.PNT1 e.PRN1 e.CURS e.ALPH ø.CHAR 1.C2 Ø.1.SZR eCURSR e.HITC BOG	JCARRY ON VECTORS FROM SCREEN JNO HIT; TRY SURFACE JRING BELL FOR HIT JIS IT "2" FOR 2ND POINT?
00667'0060525 00670'0060035 00671'002757 00672'0060075 00673'000641' 00674'000642' 00675'000521 00675'000521 00677'0060065 00700'0000007 00701'0060005 00702'000522' 00703'001000' 00704'001001' 00705'0060405 00706'020614 00707'024620 00710'106414 00707'024620 00710'106414 00711'002737 00712'0060075 00713'001000' 00714'001001' 00715'000422 00716'034461	;	JSR JMP E TO INPU JSR X Y JMP STA JSR 7 JSR CHAR XX YY JSR LDA LDA SUB# JMP JSR XX YY JSR XX YY JMP LDA	e.DISS eCURSR JT LOAD V e.HITC SURF1 2,PNT1 e.PRN1 e.CURS e.CURS e.CURS 1,C2 0,1,SZR eCURSR e.HITC BOG 3,PNT1	JCARRY ON VECTORS FROM SCREEN JNO HIT; TRY SURFACE JRING BELL FOR HIT JIS IT "2" FOR 2ND POINT? JNO, SOMETHING ELSE JHAVEN'T HIT A BLOCK JFIRST POINT BACK
00667'00600525 00670'0060035 00671'002757 00672'0060075 00673'000641' 00674'000642' 00675'000521 00675'000521 00677'0060065 00700'000007 00701'0060055 00702'000522' 00703'001000' 00704'001001' 00705'0060405 00706'020614 00707'024620 00710'106414 00711'002737 00712'0060075 00713'001000' 00714'001001' 00715'000422 00716'034461 00717'156414	; LOAD:	JSR JMP E TO INPU JSR X Y JMP STA JSR 7 JSR CHAR XX YY JSR LDA LDA LDA SUB# JMP JSR XX YY JMP	e.DISS eCURSR JT LOAD e.HITC SURF1 2,PNT1 e.PRN1 e.CURS e.CURS e.CURS e.LPH Ø,CHAR 1,C2 Ø,1,SZR eCURSR e.HITC BOG 3,PNT1 2,3,SER	JCARRY ON VECTORS FROM SCREEN JNO HIT; TRY SURFACE JRING BELL FOR HIT JIS IT "2" FOR 2ND POINT? JNO, SOMETHING ELSE HAVEN'T HIT A BLOCK JFIRST POINT BACK JCOMPARE
00667'0060035 00670'0060035 00671'002757 00672'006007S 00673'000641' 00674'000642' 00675'000521 00675'000521 00677'0060065 00700'000007 00701'0060055 00702'000522' 00703'001000' 00704'001001' 00705'0060405 00706'020614 00710'106414 00711'002737 00712'0060075 00713'001000' 00714'001001' 00715'000422 00716'034461 00717'156414	; LOAD:	JSR JMP E TO INPU JSR X Y JMP STA JSR 7 JSR CHAR XX YY JSR LDA LDA SUB# JMP JSR XX YY JSR XX YY JMP LDA	e.DISS eCURSR JT LOAD e.HITC SURF1 2,PNT1 e.PRN1 e.CURS e.CURS e.CURS e.LPH Ø,CHAR 1,C2 Ø,1,SZR eCURSR e.HITC BOG 3,PNT1 2,3,SER	JCARRY ON VECTORS FROM SCREEN JNO HIT; TRY SURFACE JRING BELL FOR HIT JIS IT "2" FOR 2ND POINT? JNO, SOMETHING ELSE HAVEN'T HIT A BLOCK JFIRST POINT BACK JCOMPARE
00667'00600525 00670'0060035 00671'002757 00672'0060075 00673'000641' 00674'000642' 00675'000521 00675'000521 00677'0060065 00700'000007 00701'0060055 00702'000522' 00703'001000' 00704'001001' 00705'0060405 00706'020614 00707'024620 00710'106414 00711'002737 00712'0060075 00713'001000' 00714'001001' 00715'000422 00716'034461 00717'156414	; LOAD:	JSR JMP E TO INPU JSR X Y JMP STA JSR 7 JSR CHAR XX YY JSR LDA LDA SUB# JMP JSR XX YY JMP LDA SUB#	e.DISS eCURSR JT LOAD V e.HITC SURF1 2,PNT1 e.PRN1 e.CURS e.CURS e.CURSR e.HITC BOG 3,PNT1 2,3,SER BOG	JCARRY ON VECTORS FROM SCREEN JNO HIT; TRY SURFACE JRING BELL FOR HIT JIS IT "2" FOR 2ND POINT? JNO, SOMETHING ELSE HAVEN'T HIT A BLOCK JFIRST POINT BACK JCOMPARE
00667'0060035 00670'0060035 00671'002757 00672'006007S 00673'000641' 00674'000642' 00675'000521 00675'000521 00677'0060065 00700'000007 00701'0060055 00702'000522' 00703'001000' 00704'001001' 00705'0060405 00706'020614 00710'106414 00711'002737 00712'0060075 00713'001000' 00714'001001' 00715'000422 00716'034461 00717'156414	; LOAD:	JSR JMP E TO INPU JSR X Y JMP STA JSR 7 JSR CHAR XX YY JSR LDA LDA SUB# JMP JSR XX YY JMP LDA SUB# JMP LDA	<pre>@.DISS @CURSR URF1 @.HITC SURF1 2,PNT1 @.PRN1 @.CURS @.ALPH Ø.CHAR 1,C2 Ø,1,SZR @CURSR @.HITC BOG 3,PNT1 2,3,SER BOG Ø,23,2</pre>	JCARRY ON VECTORS FROM SCREEN JNO HIT; TRY SURFACE JRING BELL FOR HIT JIS IT "2" FOR 2ND POINT? JNO, SOMETHING ELSE HAVEN'T HIT A BLOCK JFIRST POINT BACK JCOMPARE JANOTHER BLOCK (COINCIDENCE
00667'00600525 00671'002757 00672'0060075 00673'000641' 00674'000642' 00675'000521 00675'000521 00677'0060065 00700'000007 00701'0060055 00702'000522' 00703'001000' 00704'001001' 00704'001001' 00705'0060405 00706'020614 00710'106414 00711'002737 00712'0060075 00713'001000' 00714'001001' 00715'000422 00716'034461 00717'156414 00720'000417 00721'021023	; LOAD:	JSR JMP E TO INPU JSR X Y JMP STA JSR 7 JSR CHAR XX YY JSR LDA LDA SUB# JMP JSR XX YY JMP LDA SUB# JMP LDA	<pre>@.DISS @CURSR URF1 @.HITC SURF1 2,PNT1 @.PRN1 @.CURS @.ALPH Ø.CHAR 1,C2 Ø,1,SZR @CURSR @.HITC BOG 3,PNT1 2,3,SER BOG Ø,23,2</pre>	;CARRY ON VECTORS FROM SCREEN ;NO HIT; TRY SURFACE ;RING BELL FOR HIT ;IS IT "2" FOR 2ND POINT? ;NO, SOMETHING ELSE ;HAVEN'T HIT A BLOCK ;FIRST POINT BACK ;COMPARE ;ANOTHER BLOCK (COINCIDENCE ;HIT ON SAME BLOCK

00725'102400		SUM	6.20 -	
00726'041023		STA	0,23,2	JSET LOADS TO EERO
00727'041024		STA	0,24,2	
00730'0060115	REDR:	JSR	P.PAGE	
00731 0060525		JSR	C.TPRN	
00732'0060035		JSR	e.DISS	
00733'0060215		JSR	0.LPLS	
00734'102520		SUBEL	0.0	
09735'049000-		STA	0 LFAP	
00736'002712		J.12	PCUFSR	
00737 034440	BOG:	LDA	31-111	
007401021401		LDA	0,1,3	; XXC
00741 024437		LDA	1 . XX	JEND 2
00742'106400		SUB	0,1	FELATIVE VECTOR
00743'0300225		LDA	2. VFAC	SCALING FACTOR
80744'102400		SUB	0.0	
00745'073301		MUL		
00746'021423		LDA	0,23,3	JOLD XX LOAD
00747 040427		STA	Ø,OLDX	A PARTY LANDYS
00750'045423		STA	1,23,3	INEW XX LOAD
00751'021403		LDA	0,3,3	;YYC
00752'024427		LDA	1, YY	
00753'106400		SUB	0,1	
00754'102400		SUB	0,0	
00755'073301		MUL		
00756 021424		LDA	0,24,3	JOLD YY LOAD
00757 045424		STA	1,24,3	INEW YY LOAD
00760'024416		LDA	1.OLDX	
00761 107004		ADD	0,1,SER	; SKIP IF BOTH ZERO
00762'000746		JMP	REDR	;RE-DRAW ALL
00763'021401		LDA	0,1,3	3 XXC
00764'025403		LDA	1,3,3	JAAC
00765'0060105		JSR	e.PLTS	
00766'000000		Ø		
00767'020411		LDA	Ø, XX	
00770'024411		LDA	1.YY	
00771 0060105		JSR	e.PLTS	PLOT SINGLE NEW VECTOR
00772'000001		1		
00773'102520		SUBZL	0.0	
00774'040000-		STA	Ø. LPAP	
00775'002653		JMP	PCURSR	
00776'000000	OLDX:	0		
00777 000000	PNT1:	Ø		
01000'000000	XX:	Ø		
01001.000000	YY:	Ø		NUMBER OF PROPERTY THREE
C. Statistics	; ROUTI		NPUT OF S	SURFACE PROPERTY TYPES
01002'100257'	RET3S:	eRET3		
01003'000436'	CURSS:	CURS		
01004'000000	EIMM:	Ø		
01005.000000	DIGIT:	0		
010061000000	DIGAS:	(1		

DIGAS: 01006'000000 Ø C-129

20000 01007 020000 LBI1: "0+200 01010'000260 NØ: 01011'000271 "9+200 N9: 6 01015.000000 MOVE: START=25 SS: START+1 000025 01013'000026 01014'000027 SL: 01015'007777 TMSK: 01016'020772 SURF1:

START+2 7777 LDA 0.NO

-					
	01017131400		INC	0.0	
	01020 040766	SURF:	STA	n, DIGAS	SAVE ASCII FORM OF DIGIT
	01021 024767		LDA	1.00	
	01022'030767		LDA	2,N9	
	01023'142033		ADCE#	2, C, SNC	JCHECK FOR DIGIT O TO 9
	01024'106032		ADC2#	0,1,SEC	
	01025'000454		JMP	UTRY	;NOT DIGIT. EXIT!
	01026'122400		SUB	1,0	JBINARY VALUE
	01027 940756		STA	Ø,DIGIT	
	01030'0060365		JSR	e.HITS	FIND WHICH EDGES
	01031 000641	XRR:	х		
	01032.000642.	YRR:	Y		
	01033'002750		JMP	OCURSS	; PUT UP CURSOR AGAIN
	01034'054750		STA	3, EIMM	
	01035'010443		ISZ	FLAG	JRECORD TYPE CHANGES
		STORE	TYPE # IN	APPROP	RIATE WORD
	01036'021000		LDA	0,0,2	; CONTROL WORD
	01037'034750		LDA	3,LBIT	
	01040'117414		AND#	0,3,SER	;LONG BLOCK?
	01041 000406		JMP	LONG	
	01042'135120		MOVEL	1,3	
	01043'157000		ADD	2,3	
	01044'020747		LDA	0,55	
	01045'117000		ADD	0,3	
	01046'000406		JMP	NOSE	
	01047'135120	LONG:	MOVEL	1,3	
	01050'137000		ADD	1,3	
	01051'157000		ADD	2,3	
	01052'020742		LDA	Ø,SL	
	01053'117000		ADD	0.3	
	01054'021400	NOSE:	LDA	0.0.3	
	01055'024740		LDA	1.TMSK	
	01056'107400		AND	0,1	MASK OFF OLD TYPE #
	01057 020726		LDA	Ø,DIGIT	
	01060'103120		ADDZL	0.0	
	01061'103120		ADDEL	0,0	
	01062'101300		MOVS	0.0	; IN LEFT 4 BITS
	01063'107000		ADD	0,1	SADD IN NEW TYPE #
	01064'045400		STA	1,0,3	FUT COMPOSITE BACK
		; PRINT	DIGIT AT	CENTRE (DF EDGE
	01065 030725		LDA	2,MOVE	
	01066'022743		LDA	ØJEXRR	
	01067'142400		SUB	2,0	
	01070'026742		LDA	1, OYRR	
	01071 146400		SUB	2,1	
	01072'0060105		JSR	e.PLTS	
	01073'000000		Ø		
	01074'0060405		JSR	0.ALPH	
	01075'020711		LDA	Ø.DIGAS	
	01076'0060375		JSR	e.PRN2	
	01077 002705		JMP	0ZIMM	SRE-ENTER FOR FURTHER HITS
	01100.000000	FLAG:	Ø		

Ø,FLAG Ø,Ø,SNR LDA 01101'020777 UTRY: 01102'101005 MOV JMP PRET3S SEXIT, NO CHANGES 01103'002677 ; TO REQUEST UPDATE CYCLE. STORING ;NEW TYPE #S IN CONTACT LISTS 2. CPNT 01104'0300165 LUA 011051.011.02 0,2,2 INEN WORD LDA STA 0,00,2 01106'043000

81107 0060125		JSR	e.ALLB	DO AN UPDATE
01110 0300165		LDA	2. CPNT	
011111.051001		LDA	0,1,2	JOLD WORD
01112'043000		STA	0,00,2	
01113'002667		JMP	PRET35	JEXIT
	3			
	3	ROUTINE	TU PLOT	SINGLE BLUCK
	1			
(1114 1 11677 !		FRAC		
	RET31:	CRET35		
01116'001457'	AC2TS:	AC2SV		
AND BARRIER DE ANDRESS	VET:	VETO		
01120'001443'	PO:	POS		
	;			
01121'0060435	ONLY:	JSR	0.MESS	
01122'001474'		OMESS		
01123'177242		-350.		
01124'001274		700.		
01125'0060055	OCUR:	JSR	e.CURS	SELECT SINGLE BLOCK
01126'001452'		OCHAR		
01127'001453'		0X		
01130'001454'		OY		
01131 0060075		JSR	e.HITC	; IS IT A BLOCK
01132'001453'		OX		
01133'001454'		OY		
01134'000771		JMP	OCUR	;NO HIT RETURN
01135'052761		STA	2,0AC2T3	5 ; GOOD HIT RETURN
01136'0060115		JSR	e.PAGE	
01137 0060525		JSR	e.TPRN	The second se
01140'032756		LDA	2, 0AC2TS	
01141'0060455		JSR	e.DISB	JDISPLAY IT
01142 0060435		JSR	e.MESS	
01143'001506'		CTMES		
01144'177634		-100.		
01145'001274		700.		
01146'0060435		JSR	e.MESS	
01147'001521'		XCHES		
01150'000175		125.		
01151'001236		670.		10103 (************************************
01152'032744		LDA	2,0AC2TS	
01153'021001		LDA	0,1,2	JX CENT
01154'0060405		JSR	e.ALPH	ACCINE IT
01155'0060465		JSR	e.IPRN	FPRINT IT
01156'000005		5		the second secon
01157'032737		LDA	2,0AC2T3	
01160.051005		LDA		IXC LO PRECIS
01161'006733		JSR	OFRIC	
01162'0060435		JSR	e.MESS	
01163'001527'		YCMES		
01164'000175		125.		

01165'001212 01166'032730 01167'021003 01170'0060405 01171'0060465 01172'000005 01172'000005 01173'032723 01174'021004 01175'006717 01176'032720

650. 2,0AC2T5 LDA 0,3,2 SYCENT LDA e.ALPH JSR JSR e.IPRN **#PRINT IT** 5 2. PACETS LDA SYC LO PREC LDA 0,4,2 JSR eFRIC BLOCK POINTER 2, PACETS LDA

01177'021001	LDA	0,1,2 ;XC
01200.022003	LDA	1,3,2 JYC
01201 0060105	JSR	@.PLTS
01202.030030	Ø	
01203 021014	LDA	0,14,2 ;WEIGHT
01204 0060405	JSR	9.ALFH
01205'0060465	JSR	C.IPRN JFRINT IT
01206'000004	4	
01207 0060435	JSR	0.MESS
01210'001547'	LDMES	
01211'176504	-700.	
01212'001274	700.	
01213'0060435	JSR	0.MESS
01214'001556'	XLMES	
01215 001325	725.	
01216'091236	670.	
01217 032677	LDA	2,0AC2TS ;GET BLOCK POINTER
01220'021023	LDA	0,23,2 ;X LOAD
01221 101132	MOVEL#	0,0,SEC ;GET SIGN OF LOAD
01222 006675	JSR	OVET JPRINT "-"
01223 '006675	JSR	QPO ;PRINT "+"
01224 0060405	JSR	0.ALPH
01225 0060465	JSR	Q.IPRN ;PRINT IT
01226'000005	5	
01227 0060435	JSR	e.MESS
01230'001612'	YLMES	
01231 001325	725.	
01232'001212	650.	
01233'032663	LDA	2, EAC2TS
01234 021024	LDA	0,24,2 ; Y LOAD
01235'101132	MOVEL#	0,0,SEC ;GET SIGN OF LOAD
01236'006661	JSR	OVET
01237 006661	JSR	0PO ;PRINT +
01240'0060405	JSR	e.ALPH
01241 0060465	JSR	@.IPRN ;PRINT IT
01242'000005	5	
01243'060477	READS	0 31 VEL, FSUMS, ETC
01244'101123	MOVEL	0,0,SNC
01245'000552	JMP	OMIT
01246'0060435	JSR	0.MESS
01247 001632	XFSM	
01250'001325	725.	
01251 000702	450.	
01252'032644	LDA	2, PAC2TS ;GET BLOCK POINTER
01253'021007	LDA	0,7,2 ;XFORCE SUM
01254'101132	MOVEL#	0,0,SEC ;GET SIGN
01255'004561	JSR	VETO
01256'004565	JSR	POS
01257 0060405	JSR	e.ALPH
01260 0060465	ISR	9. IPRN

01260'0060465 01261'000006 01262'0060435 01263'001641' 01264'001325 01265'000644 01266'032630 01267'021016 01270'101132 01271'004545 01272'004551

JSR e.IPRN 6 JSR e.MESS YFSM 725. 420. 2, PAC2TS LDA 0.16.2 ;Y FORCE SUM LDA 0.0.SEC ;GET SIGN MOVEL# JSR VETO POS JSR

3		1112711		C-133
	01273 0060405	JSR	R.ALPH	
	01274'0060465	JSR	0.IPRN	
	01275.000006	6		
	01276 0069435	JSR	0.MESS	
	01277 001650	MSUM		
	01300'061325	725.		
	01301'000606	390.		
	R13021030555	LDA	2,AC25V	
	01303'021017	LUA	0,17,2 ; 10 VENT SUM	
	01314111138	10V+L#	M. D. SEC JGET SIGN	
	01305'004531	JSR	VETO	
	01306'004535	JSR	POS	
	01307 . 0060405	JSR	9.ALPH	
	01310'0060465	JSR	8.1PRN	
	01311 000007	7		
	01312 0360435	JSR	0.MESS	
	01313'001655'	XVLM		
	01314'001325	725.		
	01315'000512	330.		
	01316'030541	LDA	2,AC25V	
	01317 021005	LDA	0,5,2 ;X VELOCITY	
	01320'101132	MOVEL#	and the second sec	
	01321'004515	The second s	0,0,SEC	
		JSR	VETO	
	01322'004521	JSR	POS	
	01323'0060405	JSR	0.ALPH	
	01324'0060465	JSR	0.IPRN	
	01325'000006	6		
	01326'0060435	JSR	e.MESS	
	01327 001663	YVLM		
	01330'001325	725.		
	01331 000454	300.		
	01332'030525	LDA	2,AC2SV	
	01333'021015	LDA	0,15,2 ;Y VELOCITY	
	01334'101132	MOVEL#	0,0,SZC	
	01335'004501	JSR	VETO	
	01336'004505	JSR	POS	
	01337 '0060405	JSR	@.ALPH	
	01340'0060465	JSR	e.IPRN	
	01341 000006	6		
	01342 0060435	JSR	e.MESS	
	01343'001671'	RVLM		
	01344'001325	725.		
	01345'000416	270.		
	01346'030511	LDA	2,AC2SV	
	01347'021006	LDA	0,6,2 ;ROT VEL	
	01350'101132	MOVEL#	0,0,SEC	
	01351 004465	JSR	VETO	
	01352 004471	JSR	POS	
	01353'0060405	JSR	0.ALPH	
	01354'0060465	JSR	P.IPRN	
	01355'000006	6	half and and the	
	01356'0060435	JSR	e.MESS	
	01000 0000405	othe	a state of the sta	

01357'001535' 01360'001325 01361 000310 01362'030475 01363'021000 01364 101132 01365'004451 01366 004455

SINE 725. 200. LDA LDA MOVEL# JSR JSR POS

2,AC2SV ;GET BLOCK POINTER 0,0,2 ;SIGN OF THE SINE 0,0,SEC ;+=0,-=1 201632 VETO

					0 101
01367 021011		LDA	0,11,2	GET THE	SINE
01370'0060465		JSR	0.1PRN		
01371'177772		-6			
01372 0060435		JSR	e.MESS		
01373'001542'		DALF			
01374'901325		725.			
01375 000252		170.			
The second se			U ACOSU		
01376'030461		LDA	2,AC2SV	ACET DEL	TUETA
01377 021022		LDA	0,22,2	GET DEL	THETA
01400'040416		STA	Ø,DELF	SAVE IT	
01401 101133		MOVEL#	0.0.SNC	1- OR +	
01402'000407		JMP	LUS	JWAS POS	
01493'004433		JSR	VETO	;PRINT-	
01404'000401		JMP	• + 1	INO OP	
01405'020411		LDA	Ø,DELF		Carrier a period
01406'0060465		JSR	e.IPRN	SPRINT IT	L LESSTRATCHERD
01407 177772		-6			
01410'000407		JMP	• +7		
01411'004432	LUS:	JSR	POS	; PRINT +	
01412'020404		LDA	Ø,DELF		
01413'0060465		JSR	e.IPRN		
01414'177772		-6			
01415'000402		JMP	.+2		
01416'000000	DELF:	0			
01417 0060435	OMIT:	JSR	0.MESS		
01420'001563'		QUES			
01421 000144		100.			
01422'000144		100.			
01423'060110	DOVR:	NIOS	TTI		
01424'0060025		JSR	0.GETT		
01425'0060375		JSR	0.PRN2		
01426'024427		LDA	1.YCHAR		
01427'106405		SUB	0.1. SNR		
01430'000420		JMP	LODE		
01431 024425		LDA	1.NCHAR		
01432'106404		SUB	0,1,SER		
01433'000770		JMP	DOVR		
01434'002401		JMP	PRTST	JEXIT	
01435'101115'	RT3T:	PRETOT	entor		
01436'054422	VETO:	STA	3,AC3SV		
01437 0060065	V210.	JSR	e.PRN1		
01440 0000055			e . FRIVI		
			2 40254		
01441 034417		LDA	3.AC3SV		
01442*001401		JMP	1,3		
01443'054415	POS:	STA	3,AC3SV		
01444'0060065		JSR	0.PRN1		
01445'000053		"+			
01446'034412		LDA	3.AC3SV		
01447 001400		JMP	0,3		
01450'030407	LODE:	LDA			OCK POINTER
01451 0060535	-	JSR	e.LODE	GO TO IN	PUT ROUTINE
01452'000000	OCHAR:	0			

*IN

01452'000000 OCHAR: 3 Ø 01453 000000 OX: 01454'000000 OY: Ø my. 01455'000131 YCHAR: 01456'000116 **N NCHAR: 01457 000000 AC25V: 0 01460.000000 AC35V: Ø

01461'047111 PMESS: .TXT

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	T JO IN T PR ES SUR S* OMESS: LE CT IN GL E BL CT IN GL E BL CC K* CTMES: NT RO ID C OO RD IN AT ES XCMES: CE NT RO ID C NT SC E ST RO ID C RD IN SC E ST ST ST ST ST ST ST ST ST ST ST ST ST	T JO IN T FR ES SU RE S* OMESS: .TXT LE CT S IN GL E BL OC K* CTMES: .TXT NT RO ID C OO RD IN AT ES * XCMES: .TXT CE AT RO ID * YCMES: .TXT CE NT RO ID * YCMES: .TXT CE NT RO ID * YCMES: .TXT CE NT RO ID * YCMES: .TXT CE NT RO ID * YCMES: .TXT

01544 944124 01545 052105 01546'000101 01547 050101 01550'046120 01551'042511 01552'020104 01553'047514 01554'042101 01555'000123

ET .TXT LDMES: PL

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D LO

AD S* *AP

1556'020130	XLMES:	• TXT	*X	
1557'047514	LO	• • • •		
1560'042101	AD			
	HU I			
1561 020040	-			
1562'000000	*			
1563'047504	QUES:	•TXT	*D0	
1564'054440	Y			
1565 952517	ou			
1566'053440	W			
1567 051511	IS			
1570'020110	н			
1571 047524	то			
1572'041440	C			
1573'040510	HA			
1574'043516	NG			
1575'020105	E			
1576'044124	TH			
1577 020105	E			
1600'047514	LO			
1601'042101	AD			
1602'020123	S			
1603'020050	C			
1604'020131	Y			
1605'051117	OR			
1606'047940	N			
1697 024440)			
1610'037440	?			
1611 000040	*			
1612'020131	YLMES:	•TXT	*Y	
1613'047514	LO			
1614'042101	AD			
1615'020040				
1616 000000	*			
1617'044440	INMS:	• TXT	* I	
1620'050116	NP			
1621'052125	UT			
1622'043040	F			
1623'052454	•U			
1624'046954	12			
1625'047440	0			
1626'020122	R			
1627 020117	0			
1630'020077	?			
1631 000000	*			
1632'020130	XFSM:	•TXT	*X	
1633'047506	FO			
1634'041522	RC			
1635 020105	F			

C-13F

01640'000000 01641 020131 01642'047506 01643'041522 RC 01644'020105 E SU 01645 052523 01646'020115 M R1647'000000 * MSUM: 01650'047515 M . _ 01651 027115

01635 020105

01636'052523

01637 020115

> * YFSM: •TXT FO

E

1

SU

: .TXT *MO

*Y

01652'051440 S 01653'046525 UM 01654 000040 * 01655 920130 XVLM: • TXT *X 01656'042526 VE 01657 047514 LO 01660'044503 CI 01661'054524 TY 01662'000040 * 01663'020131 YVLM: .TXT *Y 01664 042526 VE 01665 047514 LO 01666'044503 CI Ø1667'054524 TY 01679'000040 * 01671 047522 RVLM: *RO . TXT £1672'027124 Τ. V 01673 053040 01674'046105 EL 01675'020056 . 01676'000000 * JTO PRINT FRACTION (WITH N DECIMAL (JPLACES) FOLLOWING HI PREC COORD ; NO. OF DIGITS 000004 N=4 01677 054413 FRAC: STA 3,FSAV STA Ø,FR 01700'040413 01701 0060065 0. PRN1 JSR 01702 000056 01703'024410 1.FR LDA 01704'030410 LDA 2,01000 01705 102400 SUB 0,0 01706 073301 MUL e.IPRN 01707 0060465 JSR 01710 177774 -N 01711 002401 JMP efsav 0 01712 000000 FSAV: 01713'000000 FR: Ø ; SET AT 10**N C1000: 10000. 01714'023420 . END

C-137

-	-				
			.TITL	CYCLE	
		; SEVERAL	. ADDITIC	ONAL UTILITY PROC	RAMS
			.ENT	OPTIN, .STEP, . TPF	RN
			.ENT	.KET, .RSET	
			.EXTD	. IPRN PRN1 MES	S
			.EXTD	.NVEC VFAC DIS	
				.PRN2, GETT, DBI	
			.EXTD		
			•EXTD	•M1 • • VEC • • PFLG • •	NUM
			•EXTD	.MOTFORD	
			•EXTN	CONTR	
			.ZREL .		
	00000-000123'	.RSET:	CHNGIT		
	00001-000314'	.STEP:	STEP		
	00002-000333'	.TPRN:	TPRN		
	00003-000000	.ITLO:	0		
	00004-000000	.ITHI:	0		
	00005-000000	.OPTN:	0		
	00006-000000	.COPY:	0		
	00007-0000000	.STOP:	0		
	00010-000001	.COPCT:	1		
	00011-000000	.KEFL:	0	JØ=NO KE CALC	
	00012-000011'	.KET:	KET		
	00013-000005	.C10:5			
		274.54.5	.NREL		
		1	a space of the		
		ROUTINE	TO SET	VELOCITIES TO ZE	RO
				ERGY PEAK	
	000001000000	VEET.	0		
	00000'000000	KRET:	0		
	00001 000000	POINT:	0		
	00002.000000	COUNT:	0		
	00003.000000	KHI:	0		
	00004 000000	KLO:	Ø		
	00005'000000	KOHI:	0		
	00000.900000	KOLO:	0		
	00007 '000000	FLAG:	0		
	00010.000000	HYS:	Ø	dialest realized	
		;	101100-000	-	
	00011.050011-	KET:	LDA	Ø, .KEFL	
	00012'101005		MOV	0,0,SNR	
	00013'001400		JMP	0,3	
	00014'054764		STA	3,KRET	
	00015 0340145		LDA	3. • M1	
	00016 054763		STA	3, POINT	
	00017 024764		LDA	1,KHI	
	00020 044765		STA	1,KOHI	
	00021 024763		LDA	1.KLO	
	00022'044764		STA	1,KOLO	
	00023 0240175		LDA	1 NUM	
	00024'044756		STA	1.COUNT	
	00025'102400		SUB	0.0	
	00026'040755		STA	Ø,KHI	
	00027 040755		STA	Ø,KLO	

STA Ø,KLO 00027 040755 ; TO FIND KINETIC ENERGY ITER: LDA 3. PPOINT 00030 036751 SUBEL 0.0 00031 102520 Ø,FLAG STA 00032'040755 I X VELOCITY 2,5,3 00033'031405 LDA MOVL# 2,2,520 00034'151112 BACK:

 00035'150400		NEC	0.0
		NEG	2,2
00036'145000		MOV	2,1
00037 102400		SUB	0,0
00040'073301		MUL	
00041 030742		LDA	2,KHI
00042'034742		LDA	3.KLO
00043'167022		ADDZ	3,1,SZC ; DOUBLE PREC ADD
00044'151400		INC	2,2
00045'143000		ADD	2,0
00046'040735		STA	Ø,KHI
00047 044735		STA	1,KLO
00050 014737		DSZ	FLAG
00051 000404		JMP	NEXT
	; Y VEL	OCITY	
00052 336727		LDA	3, PPOINT
00053'031415		LDA	2,15,3
00054'000760		JMP	BACK
00055 010724	NEXT:	ISE	POINT
00056'014724		DSZ	COUNT
00057'000751		JMP	ITER
	J CHECK	ON HYST	ERESIS COUNT
00060 010730		ISZ	HYS
00061 '024723		LDA	1,KLO
00062'020721		LDA	Ø,KHI
00063'030722		LDA	2,KOHI
00064'034722		LDA	3,KOLO
00065'166422		SUBE	3,1,SEC ;DOUBLE PREC SUB
00066'142401		SUB	2,0,5KP
00067 142000		ADC	2,0
00070'101123		MOVEL	0,0,SNC
00071'000431		JMP	NOPK
00072'024013-		LDA	1C10
00073'020715		LDA	Ø,HYS
00074'106032		ADCZ#	0,1,SEC
00075'000425		JMP	NOPK
	J ZERO	VELOCITI	ES
00076'0300145		LDA	2,.M1
00077'0240175		LDA	1 NUM
00100'124400		NEG	1,1
00101'102400		SUB	0,0
00102'035000	ITRE:	LDA	3,0,2
00103'041405		STA	0,5,3
00104'041406		STA	0,6,3
00105'041415		STA	0,15,3
00106'151400		INC	2,2
00107'125404		INC	1,1,SZR
00110'000772		JMP	ITRE
001111176400		SUB	3,3
00112'054676		STA	3,HYS
00113'0340165		LDA	3, PFLG JINHIBIT PRINTING IN NOPLT
00114'175004		MOV	3,3,52R
00115100010F		11453	NORK

00115'000405 JMP NOPK JSR e.MESS 00116'0060035 00117'000641' KMS 00120'001522 850. 50. 00121'000062 00122'002656 NOPK: JMP EKRET 3

J ----- RESET ROUTINE ----

1

-	00123'054407	CHNGIT:	STA	3, SAV3		
	00124'176400	011110111	SUB	3,3		
	00125 054004-		STA	3 ITHI		
	00126'054003-		STA	3 ITLO		
	00127 176520		SUBEL	3,3		
	00130'054010-		STA	3COPC	T	
			JMP	eSAV3	•	
	00131'002401	SAV3:	Ø	eshvs		
	00132'000000	SAVS.	6			
			OPTION	INPUT R	OUTINE	
			- UPIION	INPUT N	OUTINE	
	00100100/0075	ODTINA	100	A PACE		
	00133'0060075	OPTIN:	JSR	e.PAGE		
	00134'0060035		JSR	e-MESS		
	00135'000455'		OPTMS			
	00136'177242		-350.			
	00137 001274		700.			
	00140'0060035		JSR	e.MESS		
	00141 000467		CRMS			
	00142.000062		50.			
	00143'001236		670.			
	00144'0060115	OUT:	JSR	e.GETT		
	00145'024546		LDA	1, CRGRT	ANTER EVIT	
	00146'106415		SUB#		IMUST EXIT	
	00147 000535		JMP	HOME		
	00150'0060035		JSR	e.MESS		
	00151 000523		N1			
	00152'000310		200.			
	00153'001212		650.	0 4555		
	00154'0060035		JSR	e.MESS		
	00155'000555'		01			
	00156'000113		75.			
	00157'001130	041.	600. JSR	e.GETT		
	00160'0060115 00161'024531	0.11	LDA	1.YCHR		
	00162'106414		SUB#	0,1,SZR		
	00163'000405		JMP	.+5		
	00164'0060105		JSR	e.PRN2	JPRINT Y	
	00165'126520		SUBEL	1,1		
	00166'0440045		STA		SET FLAG TO	PRINT
	00167'000407		JMP	CNT1	INEXT	
	00170'024521		LDA	1.NCHR	CHK FOR NO	
	00171'106414		SUB#	Ø,1,SER	Form For no	
	00172'000766		JMP	OV1		
	00173'0060105		JSR	e.PRN2	PRINT IT	
	00174'126440		SUBO	1.1	STREET II	
	00175'0440045		STA	The second second second second second	INHIBIT PRIM	TING
	00176'0060035	CNT1 .	JSR	8.MESS	FINIDIT FRI	
	00177'000605'	01111	02	0.11233		
	00200'000113		75.			
	00200 000113		550.			
	00202 0060125		JSR	e.DBIN		
	00202 0000125		STA		ISET SCALE FO	OCT

00203'0440055 00204'0060035 00205'001051' 00206'000113 00207'000764 00210'0060115 OVR6: 00211'024501 00212'106414 00213'000405

STA JSR Q6 75. 500. JSR LDA SUB# JMP

1..VFAC ISET SCALE FACT 0.MESS

0.GETT 1.YCHR 0.1.SZR .+5

00214'0060105		JSR	e.PRN2	JPRINT Y
00215'126520		SUBEL	1,1	
00216'044011-		STA		SET FLG TO K.E. ZERO
00217'000407		JMP	CTNU	INEXT
00220'024471		LDA	1.NCHR	
00221'106414		SUB#	0,1,SZR	
00222'000766		JMP	OVR6	
00223'0060105		JSR	e.PRN2	
00224'126440		SUBO	1 . 1	
00225'044011-		STA	1. KEFL	JINHIB K.E.ZERO
00226'0060035	CTNU:	JSR	e.MESS	
00227'000646'		03		
00230'000113		75.		
00231'000702		450.		
00232'0060115	0V2:	JSR	e.GETT	
00233'024456		LDA	1,NCHR	
00234'106414		SUB#	Ø,1,SER	
00235'000405		JMP	.+5	
00236'0060105		JSR	0.PRN2	JPRINT N
00237'126440		SUBO	1,1	
00240'044005-		STA	1 OPTN	JNO OPTIONS
00241'000433		JMP	LAST	
00242'024450		LDA	1.YCHR	
00243'106414		SUB#	0,1,SZR	
00244'000766		JMP	0V2	
00245'0060105		JSR	e.PRN2	JPRINT Y
00246 126520		SUBZL	1 . 1	
00247 044005-		STA	1. OPTN	JSET OPTION FLAG
00250'0060035		JSR	e.MESS	
00251'000756'		N2		
00252'000144		100.		
00253'000620		400.		
00254'0060035		JSR	e.MESS	
00255'001010'		N3		
00256'000175		125.		
00257 000567		375.		
00260'0060035		JSR	e.MESS	
00261'000676'		04		
00262'000113		75.		
00263'000505		325.		
00264'0060125		JSR	e.DBIN	
00265'044006-		STA	1. COPY	
00266'0060035		JSR	e.MESS	
00267 000727		05		
00270'000113		75.		
00271'000423		275.	-	
00272'0060125		JSR	e.DBIN	
00273'044007-		STA	1.STOP	
00274'0060035	LAST:	JSR	e.MESS	
00275'001033'		N4		
00276'000310		200.		

۰.

00277 000257 175. JSR 00300'0060115 OV3: 00301'024412 LDA 00302'106414 SU8# 00303'000775 JMP JSR 00304'0060075 HOME: JSR 00305'006002-00306'0060065 JSR 00307 '002401 JMP

e.GETT 1,CRGRT 0,1,SER OV3 e.PAGE e.TPRN e.DISS eBAKK

				L-1
00310'177777	BAKK:	CONTR		
00311'000116	NCHR:	"N		
00312'000131	YCHR:	ny		
00313'000015	CRGRT:	15		
00313-000013	CROAL.	15		
	,	ROUT	INE TO S	TEP CYCLE COUNTER
			THE TO D	TET GTOLE GOOMTEN
	:	JSR	P.STEP	
	1	Jak	6.DIEL	
00314'054523	STEP:	STA	3, SAV3P	
	SILF.	LDA	Ø. ITLO	
00315'020003-				
00316'024514		LDA	1.ITMAX	
00317'101400		INC	0.0	
00320'106415		SUB#	0,1, SNR	
00321 000404		JMP	NOTCH	
00322'040003-		STA	Ø ITLO	
00323'034514		LDA	3, SAV3P	PARENT OF THE
00324'001400		JMP	0.3	JEXIT
00325'102400	NOTCH:	SUB	0.0	
00326'040003-		STA	Ø,.ITLO	;RESET LO WORD
00327 010004-		ISZ	.ITHI	JINCREMENT HI WORD
00330'004434		JSR	OPTON	CHECK OPTIONS
00331 034506		LDA	3, SAV3P	
00332'001400		JMP	0.3	JEXIT
	1			
	1	ROUT	INE TO PI	RINT CYCLES
	3			
	;	JSR	e.TPRN	
	1			
00333'054501	TPRN:	STA	3. TERMI	TE
00334'060477		READS	Ø	
00335'101222		MOVER	0,0,SEC	
00336 000425		JMP	OOT	
00337 '0060035		JSR	e.MESS	
00340'000454'		MAT		
00341 000702		450.		
00342'001402		770.		
00343'020004-		LDA	Ø. ITHI	
00344'0060015		JSR	e.IPRN	HI PART
00345'000005		5	4.5.61	
00346'020003-		LDA	Ø. ITLO	
00347 0060015		JSR	8.IPRN	JLO PART
00350'177774		-4		EADING ZEROS
00351'0060035		JSR	e.MESS	
00352'000440'		CYC		
00353'001116		590.		
00354'001402		770.		
00355'0240135		LDA	1.MU	
00355'0240135		LDA	2,01000	
		SUB	0,0	
00357'102400 00360'073301		MUL	010	
00360 073301		JSR	R. TPRM	PRINT DEFAULT MU
		and the last		

00361'0060015 JSR 0.IPRN JPRINT DEFAULT MO 00362'177775 -3 00363'002451 OOT: JMP @TERMITE J

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OPTION CHECKER

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-					
	00364'054452	OPTON:	STA	3, SAVE3	
	00365'020005-		LDA	Ø. OPTN	JACTIVATE OPTIONS ?
	00366'101005		MOV	0,0, SNR	THAT THE FREE TRACES
	00367 001400		JMP	0,3	
	00370 020006-		LDA	Ø. COPY	
	00371'101004		MOV	0,0,SER	
	00372'004413		JSR	COPI	
	00373 020007-		LDA	Ø. STOP	
	00374'101004		MOV	0,0,52R	
	00375'000403		JMP	BON	
	00376'034440		LDA	3, SAVES	
	00377 001400		10000	and the second second	
		DONIA	JMP	0,3	
	00400'024004-	BON:	LDA	1ITHI	
	00401'106405		SUB	0,1, SNR	
	00402'002431		JMP	econtin	
	00403'034433		LDA	3, SAVE3	
	00404'001400		JMP	0,3	
		;			
	00405'054430	COPI:	STA	3, SAV3A	
	00406'020004-		LDA	Ø. ITHI	
	00407'024010-		LDA	1,.COPC	T A TELEBRELISPAG
	00410'106414		SUB#	0,1,52R	
	00411'001400		JMP	0,3	
	00412'0060025		JSR	e.PRN1	
	00413'000007		7		JRING BELL
	00414'004717		JSR	TPRN	
	00415'0060065		JSR	e.DISS	
	00416'0060025		JSR	e.PRN1	
	00417'000033		27.		JASCII ESC
	00420 0060025		JSR	e.PRN1	
	00421 000027		23.		JASCII ETB
	00422 0060075		JSR	.PAGE	NOT THE PARTY A SHEAR
	00423'024010-		LDA	1. COPC	r it sorehorteen
	00424'030006-		LDA	2. COPY	LAN LAND SIDE OF DISADS
	00425'147000		ADD	2,1	
	00426'044010-		STA	1. COPC	T 36 100000-00000
	00427 034406		LDA	3. SAV3A	I Dr. CZr-OND'T GEND
	00430'001400		JMP	0,3	
	00400 001400	3	0	070	
	00431 001750	and the second se	1000.		
	00432'023420	ITMAX:	10000.		
	00433'000310'	CONTIN:	CONTR		
	00434'000000	TERMITE:			
	00435 000000	SAV3A:	0		
	00436'000000	SAVE3:	0		
		SAVES: SAV3P:	0		
	00437'000000		.TXT	* C	
	00440'041440	CYC:			
	00441'041531	YC			
	00442'042514	LE			
	00443'020123	S			

00444'020040 00445'042504 DE 00446'040506 FA 00447'046125 UL 00450 020124 T 00451 052515 MU =Ø 00452'030075 .* 00453'000056 * * .TXT 00454 000040 MAT: OPTMS: * A .TXT 00455'040440

456'040526	VA				
457 046111	IL				
460 041101	AB				
461 042514	LE				
462 047440	0				
463'052120	PT				
464'047511	10				
465 051516	NS				
466'000040	*				
467 '020050	CRMS:	.TXT	*(
470'044510	HI				
471 920124	T				
472'027103	C.				
473'027122	R.				
474'052040	т				
475'020117	0				
476'047507	GO				
477 041040	В				
500'041501	AC				
501 020113	K				
502'047516	NO				
503'020127	W				
504'020055	-				
505'047101	AN				
506'020131	Y				
507 052117	OT				
510.042510	HE				
511'020122	R				
512'042513	KE				
513'020131	Y				
514'047524					
515'041440					
516'047117					
517 044524					
520'052516	NU				
521'020105	0.1.0122				
522'000051)*				
523'040450	N1:	.TXT	*(A		
524'051516	NS				
525'042527	WE				
526'020122	R				
527'046101	AL				
530'020114					
531 052521	QU				
532'051505	ES				
533'044524	TI				
534'047117					
535'026523	S-				
536'052123	ST				
537 047101					
540 040504	Laboration of the second se				
541'042122	RD				
542'040440	A				

00542'040440 00543'051516 00544'042527 00545'051522 00546'047072 00547'031454 00550'041450 00551'024522

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NS

WE

RS

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Ø	0552	.047	054	• N	
0	0553	1047	054	,N	
Ø	0554	.000	051)*	
Ø	0555	.047	504	Q1:	
Ø	0556	.054	440	Y	
1953	0.9-10-22.01	1052		OU	
		.053			
		1051	and the second second	IS	
		1020		H	
		1047	Standard Parts	то	
		1050	Concernant State		
		1044		RI	
200		1052	0.00010000	NT	
1000		1040	COLUMN TO A	A	
-		1050	Construction of the second	PP	
26	753575140.CO	.044		LI	
1.7.8		1042		ED	
		1046		L	
27.20	and a start of the	1040		OA	
		.020		D	
124	2000 BB	1040		VA	
202	101220 (2019	.052		LU	
	30.0	1051		ES	
		1024	ALC: NOT BEEN		
1220	Sector States of the	1027		Y/	
		1024			
1000		.000		?*	
		1044		02:	
		1052		AT	
		1053		Ŵ	
		1052		ou	
		1042		LD	
2.2		.054			
775	22.17 State 1.00	.052		OU	
Ø	0614	.046	040	L	
- 222	25	.045		IK	
Ø	0616	.050	105	Ε	
Ø	0617	'051	501	AS	
Ø	0620	.052	040	Т	
Ø	0621	'042	510	HE	
Ø	0622	.053	040	V	
Ø	0623	.041	505	EC	
Ø	0624	'047	524	TO	
Ø	0625	.050	122	R	
Ø	0626	.041	523	SC	
Ø	0627	.046	101	AL	
Ø	0630	1020	105	E	
Ø	0631	.040	506	FA	
Ø	0632	1052	103	CT	
Ø	0633	1051	117	OR	
Ø	0634	1024	040	(
0	0635	.059	116		
Ø		'051			
0	9627	1027	451	12	

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.TXT

*D0

00637 037451)?
00640'000000	*
00641'027113	KMS:
00642'027105	Ε.
00643'042520	PE
00644'045501	AK
00645'000000	*

•TXT *K•

-	-	-	

00646'047504	03:
00647 054440	Y
00650'052517	où
00651 053440	W
00652 051511	IS
00653'020110	н
00654 047524	TO
00655 052440	U
00656'042523	SE
00657 040440	A
00660 052125	UT
00661 041517	OC
00662 050117	OP
00663'020131	Y
00664'051117	OR
00665'040440	A
00666'052125	UT
Contraction of the second second second	100 M (100 M)
00667'051517	OS
00670'047524	TO
00671 '020120	P
00672 054450	(Y
00673 047057	1N
00674 037451)?
00675 000000	*
00676'044127	04:
	AT
00677'052101	Contraction of the second
00700'053440	W
00701 052517	ou
00702'042114	LD
00703'054440	Y
00704'052517	OU
00705'046040	L
00706'045511	IK
00707 020105	E
00710'051501	AS
00711'052040	Т
00712'042510	HE
00713'041440	C
00714 050117	OP
00715 020131	Y
00716'047111	IN
00717'051103	CR
00720'046505	EM
	EN
00721'047105	
00722'020124	T
00723'047050	CN
00724'041454	۶C
00725'024522	R)
00726'000077	?*
00727'052101	05:
00730 053440	W
00731'040510	HA
00101 040010	-

.TXT

-TXT *DO

• TXT

*WH

*AT

00732'020124 T PO 00733'047520 00734'047111 IN 00735'020124 T 00736'047527 WO 00737'046125 UL 00740'020104 D 00741 '047531 YO

7	1.77	
	00742 020125	U
	00743'044514	LI
	00744'042513	KE
	00745'052040	T
	00746'020117	0
	00747 052123	ST
	00750'050117	OP
	00751 024040	C
	00752'026116	Na
	00753'051103	
		CR
	00754'037451)?
	00755'000000	*
	00756'047516	N2:
	00757 042524	TE
	00760 020072	:
	00761 044124	TH
	00762'020105	E
	00763'047596	FO
	00764'046114	LL
	00765'053517	OW
	00766'047111	IN
	00767 020107	G
	00770 052516	NU
	00771 041115	MB
	00772'051105	ER
		S
	00774'051101	AR
	00775'020105	E
	00776 052515	MU
	00777'052114	LT
	01000'050111	IP
	01001'042514	LE
	01002'020123	S
	01003'043117	OF
	01004'030440	1
	01005'030060	00
	01006'030060	00
	01007 000000	*
	01010'044450	N3:
	01011'026105	E,
	01012'044124	TH
		E
	01013'020105	co
	01014'047503	
	01015'050115	MP
	01016'020056	-
	01017'047111	IN
	01020'042524	TE
	01021'050122	RP
	01022'042522	RE
	01023'051524	TS
	01024'031040	2
	01025'040440	A
	01026'020123	S
	01027 030062	20
	010301030060	00

XT	*N0	

•TXT *(1

*HI

• 72

the paper of the paper want of a distance

01030'030060 00 01031'024460 0) 01032'000000 * 01033'044510 N4: .TXT 01034'020124 T 01035'040503 CA

C-148

01036'051122	RR
01037 040511	IA
01040'042507	GE
01041'051040	R
01042'052105	ET
01043'051125	UR
01044'020116	N
01045'047524	TO
01046'042440	E
01047'044530	XI
01050'000124	T*
01051'047504	06:
01052'054440	Y
01053'052517	OU
01054'053440	W
01055'051511	IS
01056'020110	н
01057 047524	TO
01060.022440	U
01061 042523	SE
01062'045440	к
01063'042456	• E
01064'055056	• 2
01065'051105	ER
01066'024117	00
01067 027531	Y/
01070 024516	ND
01071 000077	?*

• TXT *D0

. END

		20122-00-002010 0020-00-00000 0022-0-000000 0022-0-000000	

.TITL INPUT

SEVERAL INPUT ROUTINES

3

3

		.ENT	·SPRP, · INP, · UINP, · UD, · UN, · PSEG
		.ENT	FEET, POUND, MOVEL, . PEMI, . PRES
		.ENT	·LODE MOVE XCGD YCGD
		.ENT	·SYCL MFLG DMBN DMBP
		.EXTD	·PRN1, ·PLIS, ·PAGE, ·MESS, · 1PRN
		.EXID	MU, DISS, CURS, ALPH, PHN2
		•EXID	.AXIS, .DBIN, .GEII, .PRN2
		•EXTD	. IFRN, . HIIC
		.EXTD	.CHEK, . WORD, .HITS, .DB0, .M7, .MEM
		•EXID	. MSKR, . LENG, . PON1, . PON2, . REBE
		.EXTN	CONIR
		.ZREL	
00000-000277'	.SPRP:	PROP	
00001-0000000'	.INP:	INPUT	
00002-001003'	.LODE:	LODE	
00003-001157'	.SIGN:	SGN	
00004-001174'	BRNG:	BRNG	
00005-001202'	.NGAT:	NGAT	
00006-001043'	.MOVE :	MOVE	
00007-000000	.XCGD:	0	JX DISP
00010-000000	.YCGD:	0	IY DISP
00011-000000	.SYCL:	Ø	JDCM CYCLES
00012-000000	.MFLG:	Ø	JDCM FLAG - Ø=OFF
00013-009000	.DMBN:	Ø	J " BLOCK NO.
00014-000000	.DMBP:	0	J " BLOCK POINTER
20015-000000	.UD:	Ø	JUNIT OF DISPLACEMENT
00016-000000	.U. :	0	JUNIT WEIGHT
00017-000312'	.UINP:	UINP	JENIRY FOR UNITS INPUT ROUTINE
00020-177777	.PEMT:	177777	PRESS. SEGMENT EMPTY HEAD
00021-177777	.PRES:	177777	PRESS. SEGMENT LIST HEAD
00022-000413'	.PSEC:	EGGI	The Alignet State of the Alignet State of the Points
		.NREL	
000012		.RDX	10
000012	1		
	IDISPLA	Y PROPER	TY TABLE AND WAIT FOR
		Contraction of the second seco	N NEW FRICIION COEFFICIENTS.
	1		
00000 054467	INPUT:	STA	3. SPSAV
00001 '0060035	IN2:	JSR	P.PAGE
00002 0060045	1146 +	JSR	e.MESS
		TEXTI	6.4623
00003'001222'			
00004 177634		-100	
00005'001130		600	
00006 0060045		JSR	e.MESS
00007 '001234'		TEXT2	
00010'177634		-100	
00011 001034		540	
00012.0000042		JSR	e.MESS
00013 001237		TEXTS	

00013'001237' 00014'177160 00015'001034 00016'0060045 00017'001244' 00020'000144 00021'000776 00022'0200065 TEXT3 -400 540 JSR 0.MESS TEXI4 100 510 LDA 0.MU

-			100	CDAO
	000231004456		JSR	FRAC
	00054,000050		100	
	00025 000776		510	
		;INITIAI	THE REAL PROPERTY OF	PVARIABLES
	00056,030066-		LDA	2,.SPRP
	00027 151400		INC	2,2
	00030'050440		STA	2, POINT
	00031 020440		LDA	0.N16
	00032 040434		STA	Ø, CNI
	00033 014433		DSZ	CNI
	00034'102520		SUBZL	0.0 ; STARI @ 1 NOI 0
	00035 040435		STA	0.NUM
	000361020436		LDA	0.Y1
	00037 040405		STA	0, YY
	00040'040413		STA	0, 111
		SCAN TI	HROUGH PI	ROPERIY IYPES,
				ION FOR EACH
	00041 0060045	TOP:	JSR	e.MESS
	00042'001256'		TEXT5	A REAL AND THE FLAM THE CARAGE
	00043'000144		100	
	00044 000000	YY:	0	
	00045 020425		LDA	0.NUM
	00046'0060055		JSR	0.IPRN
	00047 .000002		2	
	00050'022420		LDA	Ø. POINT SPROPERTY #
	00051 '004430		JSR	FRAC
	00052 000620		400	TING
	00053 000000	YYY:	0	
	000010	111.	.RDX	8
	00054'010414		ISE	POINT
	00055'010415		ISE	NUM
	00056'020415		LDA	ØTINC
	00057 024774		LDA	1. YYY
	00060 106400		SUB	0.1 INEW Y
	00061 '044772		STA	1. YYY
	00062 044762		STA	1. YY
	00063 014403		DSZ	CNT
	P0064'000755		JMP	TOP
	00065'000446		JMP	GET
	00066 000000	CNT:	Ø	
	00067 '000000	SPSAV:	Ø	
	00070.000000	POINT:	0	
	00071 000012	N16:	12	SIZE OF PROPERTY TABLE
	00072.000000	NUM:	0	VOICE OF THOSE THOSE
	P00012 0000000	1.0.1.	.RDX	10
	000026	YRON=22	+ NUA	10 TO LOT A CARDON
		YTOP=48		
	000750			I O P
	000414		0+YROW+Y1	JDISTANCE BETWEEN LINES
	00073'000026	YINC:	YROW	
	00074'000722	Y1:	YTOP-YRO	A SAC SAC SCORE SCORES
	00075'000764	X1:	500	
	00076'000414	YL:	YBOT - BDX	0

000010 .RDX 8 00077'000215 CR: 15+200 00100'000256 DOT: ".+200 JTO PRINT FRACTION (WITH N DECIMAL JPLACES) AT (X,Y) ON SCREEN

3

3

;

JSR FRAC

-					
		1	Y		
		IFRACTIO	IN IN A	ca	
		1		40	
	000003	N=3			
	00101 054424	FRAC:	STA	3.FSAV	
	00102 946424	LUMO.	STA	ØFR	
				0,0,3	
	00103'021400		LDA		
	00104'025401		LDA	1,1,3	
	00105'0060025		JSR	e.PLTS	
	00106'000000		0		
	00107 0060015		JSR	e.PRN1	
	00110'000037		37		
	001111 0060015		JSR	P.PRN1	
	00115.000000		"0		
	00113'0060015		JSR	P.PRN1	
	00114 000056		·· .		
	00115'024411		LDA	1,FR	
	00116'030414		LDA	2,01000	
	00117'102400		SUR	0.0	
	00120 073301		MUL		
	00121 0060055		JSR	e.IPRN	
	00122'177775		-N		
	00123 034402		LDA	3,FSAV	
	00124'001402		JMP	2,3	
	00125 000000	FSAV:	Ø		
	00125'000000	FR:	0		
	00127 '000000	CHAR:	Ø		
	00130.000000	X:	Ø		
	00131 '000000	Y:	0		
	000012		.RDX	10	
	00132'001750	C1000:	1000	SET AT	10**N
	000010	01000	.RDX	8	
	000010	1			
		PUT UP	CURSOR	AND WAIT	
		1	001.001.		
	00133'0060105	GET .	JSR	e.CURS	
	00134'000127'	021.	CHAR	0.00000	
	00135'000130'		X		
	00136'000131'		Ŷ		
	00137 0060115		JSR	e.ALPH	
	00140 020767		LDA	Ø,CHAR	
	00141 024736		LDA	1.CR	
			SUB#		CHECK FOR "RETURN"
	00142'106414				CHECK FOR ACTORN
	00143'000405		JMP	NEXT	INO CHANGES RETURN.
	00144'0060035		JSR	e.PAGE	IND CHANGES RETORNS
	00145'0060165		JSR	e.TPRN	Lange starts and sold shares
	00146'0060075		JSR	e.DISS	JAND EXIT
	00147 002720	105005000000	JMP	espsav	
	00150'024730	NEXT:	LDA	1,DOT	
	00151'106414		SUB#	Ø,1,SZR	JCHECK FOR DEC. POINT
	00152'000761		JMP	GEI	INO GOOD; KEEP WAITING
	00153'024756		LDA	1.1	
				0. 11	

C-151

00154'020722 00155'106423 00156'000755 00157'102400 00160'030713 00161'073101 00162'020707 00163'122423 LDA Ø,YL SUBZ Ø,1,SNC ;CHECK FOR LOWER LIMIT JMP GET SUB Ø,Ø LDA 2,YINC DIV LDA Ø,N16 SUBZ 1,Ø,SNC ;CHECK FOR UPPER LIMIT

				C-152
00164 000424		JMP	TRYMU	
00165 030000-		LDA	2. SPRP	
00166'113000		ADD	0,2	POINTER TO PROP TABLE
00167 050437		STA	2.PPNT	
10101 030431	SET UP	A Construction of the second s	100000000000000000000000000000000000000	NT NEW NUMBER
00170 102400	JULI OI	SUB	0,0	
00171 030702		LDA	2,YINC	
		MUL	ENTING	
00172'073301		LDA	Ø,YL	
00173 020703			and the second se	
00174'107000		ADD	0,1	
00175'020700		LDA	Ø,X1 e.PLTS	
00176'0060025		JSR Ø	e.FLIS	
00177'000000		JSR	P.ALPH	
00200'0060115		LDA	Ø, CHAR	
00201'020726		JSR	e.PRN2	
00202 0060125		JSR	KEYB	
00203'004430		Contraction .	Ø, SUM	
00204'020425		LDA	2, PPNT	
00205'030421		STA	0,0,2	STORE NEW FRICTION
00206'041000		A MARTINE DE	GEI	STORE WER FRIGITOR
80207 '000724	TOWNER	JMP	I REAL TO A COMPANY AND A COMPANY	OUTON FOR DEFAULT HALLE
002101101404	TRYMU:	INC		CHECK FOR DEFAULT VALUE
00211 000722		JMP	GET	
00212'024413		LDA	1, YMU	
00213.050665		LDA	Ø,×1	
00214'0060025		JSR	e.PLTS	
00215 000000		Ø		
00216 0060115		JSR	8.ALPH	
00217 020710		LDA	Ø, CHAR	; SEND OUT DEC. POINT
00550,0020152		JSR	e.FRN2	
00221 004412		JSR	KEYB	
00222 020407		LDA	Ø, SUM	
00223 0400065		STA	Ø,MU	
00224.000707		JMP	GET	
00225'000776	YMU:	13*YROW	+YBOT	
00226'000000	PPNT:	0		
00227 .000000	NN:	Ø		
00230 000005	NTIM:	5		
00231 '000000	SUM:	0		
00232.000000	KSAV:	Ø		
00233'054777	KEYB:	STA	3.KSAV	
00234'034434		LDA	3,TBL	
00235 054432		STA	3.TBLSV	
00236'102400		SUB	0.0	
00237 040772		STA	0.SUM	
00240 020770		LDA	Ø,NTIM	
00241 '040766		STA	ØINN	
A CONTRACT OF A	CIT.	JSR		
00242'0060155	011:		e.GETI	
00243 0060125		JSR	e.PRN2	
00244 0060205		JSR	e.CHEK	
00245'000415		JMP	ERROR	
00246'105000		MOV	0,1 3.TRI SV	

00250'034420 00250'031400 00251'102400 00252'073301 00253'020756 00254'123020 00255'040754 00255'040754 LDA SUB MUL LDA ADD STA ISZ

TBLSV

2,0,3 ;GET MULTIPLIER 0,0

0,SUM 1.0 JADD IN NEW DICII 0,SUM C 152

A 46 14

DSE 00257 014750 NN 00260 000762 JMP GIT JEXIT FOR TOO MANY DIGITS 00261 002751 JMP eksav 1, CRNP 00262 024414 ERROR: LDA SUB# PP263'122415 1,0,5NR JMP ; GOOD EXIT 00264 002746 eksav BAD EXIT 00265 002401 JMP PINP 00266'000001' INP: IN2 00267 000000 TBLSV: Ø A1=77777/5 014631 000012 . RDX 10 A2=A1/10 001217 000101 A3=A2/10 000006 A4=A3/10 A5=A4/10 000000 . RDX 000010 8 00270'000271' TBL: .+1 00271 014631 AI 00272'001217 A2 00273 000101 A3 00274'000006 A4 00275 000000 A5 JCARRIAGE RET. NO PAR. 00276 000015 15 CRNP: PROP: 000000 **JTABLE FOR FRICTION COEFFICIENTS** 000012 .BLK 12 3 3 FROUTINE TO ACCEPT INPUT OF UNITS FROM SCREEN 3 . RDX 000012 10 USAV: 00311 000000 Ø 00312 054777 STA 3,USAV UINP: JSR e.PAGE 00313'0060035 00314 0060045 JSR e.MESS 00315'001264' **TEXI8** -100 00316'177634 00317 001130 600 e.MESS 00320 0060045 JSR 00321 '001305' TEXT9 00322'177634 -100 565 00323 001065 JSR e.MESS 00324 0060045 00325 001312 TEX10 226 00326'000342 00327 001065 565 e.AXIS JSR 00330 0060135 778 00331 001412 00332'000144 100 00333'000550 360 e.MESS 00334 0060045 JSR

00335'001337' 00336'000144 00337'000620 00340'0060145 00341'044015-00342'0060215 00343'000361' 00344'0060045 00345'001365'

TEX11 100 400 JGET DISTANCE UNIT JSR e.DBIN STA 1,.UD JGET STRING e.kORD JSR STORAGE LOCATION FEET e.MESS JSR TEX12

00346 000144		100		
00347 000310		200		
000010		.RDX	8	
00350 0060145		JSR	P.DBIN	JGET UNIT WEIGHT
00351 044016-		STA	1 UW	
00352 0060215		JSR	P. HORD	FORCE DESCRIPTOR
00353'000372'		POUND		ABD C.
00354'0060155		JSR	0.GETT	
00355 0060035		JSR	.PAGE	
00356'0060165		JSR	e. TPRN	
00357 0060075		JSR	e.DISS	
00360 002731		JMP	OUSAV	
000011	FEET:	.BLK	11	BYTE STRING FOR DISPL.
000011	POUND:	.BLK	11	BYTE STRING FOR FORCE
000011	1			
	INPUT	OF PRESS	URE SEGM	ENTS
	;			KOW . DOLLARS
00403 0060045	ERR:	JSR	e.MESS	
000012		. RDX	10	
00404'001417'		TOBIG		
00405'000310		200		
00406'000764		500		
000010		. RDX	8	
00407 000405		JMP	EGGS	
00410'000000	EGG3:	Ø		
00411 '000000	FORIN:	Ø		
000012		.RDX	10	
00412'000175	N125:	125		
000010		.RDX	8	
00413'054775	EGG1:	STA	3,EGG3	
00414'0060105	EGGS:	JSR	e.CURS	
00415'000604'		CHAR1		
00416'000605'		XP		
00417'000606'		YP		
00420'020564		LDA	Ø, CHARI	
00421 0060205		JSR	e . CHEK	The second
00422'002766		JMP	eEGG3	;EXIT
00423'0060115		JSR	e-ALPH	
00424'0060225		JSR	e.HITS	
00425'000605'		XP		
00426'000606'		YP	FREE	INO WIT
00427 000765		JMP	EGGS	JNO HIT JBLOCK POINTER
00430'050557		STA	2,AC2B	JEDGE #
00431 044557		STA	1 JNP	BLOCK #
00432'040557 00433'054557		STA	Ø,NB 3,ZIMM	RE-ENTRY ADDRESS
00433 054551		LDA	0.XP	THE ENTRY HOURESS
00434 020551		LDA	1,YP	
00436'030555		LDA	2,05	OFFSET
00437 142400		SUB	2,0	JULTULI
00440 146400		SUB	2,1	
00440 146400		JSR	e.PLTS	
00441 0000025		a	erreis	

00442'000000 00443'0060115 00444'0060015 00445'000052 00446'020536 00447'0060235 00450'030572 00451'142414

• • >

0 JSR JSR LDA JSR LDA SUB#

@.ALPH @.PRN1 JPRINT * ON SELECTED JEDGE 0.CHAR1 JGET INITIAL CHARACTER BACK @.DB0 JNOW GET THE REST 2.CRR 2.0.SER JCHECK FOR CR

00452'002736		JMP	BEGG3	JEXIT	
00453 044736		STA	1.FORIN		
00454'030533		LDA	2,AC28		
00455 024533		LDA	LINP		
00456'0060275		JSR	P.LENG		
00457 105000		MOV	0.1		
00460'030731		LDA	2,FORIN		
00461 102400		SUB	8.9		
00462'073301		MUL			
00463'030727		LDA	2,N125		
00464'142513		SUBL *		JCHECK BEFORE DIVIDING	
00465'000716		JMP	ERR		
00466'073101		DIV			
00467 044554		STA	1,FORCE		
00470'000572		JMP	COMPM	COMPUTE MOMENT	
00471 004440	TWIT:	JSR	EXIST	ISEE IF SEGMENT EXISTS	
00472'000463		JMP	NEWEN	INO, MAKE A NEW ONE	
00473 020550		LDA	Ø,FORCE		
00474'101004		MOV		CHECK FOR ZERO FORCE	
00475 000524		JMP	REST1	JENTER NEW FORCE IN OLD SE	G.
hears brinsta	THE FOL		Alexandra and a second s	A DEAD PRESSURE SEGMENT	
00476'021002	FILL FOR	LDA	0,2,2		
			175 C		
00477'041400		STA	0,0,3	STORE IN PREVIOUS ONE	
00500'020020-		LDA	Ø. PEMT		
00501 050020-		STA	2. PEMT		
00502'041002	Grand States	STA	0,2,2	ILINK UP WITH OTHERS	
				NY MORE HITS	
00503'034507	AGAIN:	LDA	3,ZIMM		
00504'005401		JSR	1,3	IRE-ENTER "HITS" WITH	
00505'000605'		XP		IRETURN TO HERE	
00506'000606'		YP			
00507 000705		JMP	EGGS	INO MORE HITS	
00510.054502		STA	3,ZIMM		
00511 050476		STA	2,AC2B		
00512'044476		STA	1.NP		
00513'040476		STA	Ø.NB		
00514'0060275	j.	JSR	0.LENG		
00515'105000		MOV	0,1		
00516'030673		LDA	2.FORIN		
00517'102400		SUB	0,0		
00520'073301		MUL	0,0		
00521 030671		LDA	2,N125		
00522'142513		SUBL#		CHECK BEFORE DIVIDING	
00523'000660		JMP	ERR	JUNEON DEFONE DEFENSE	
00524'073101		DIV	LINN		
00525'044516		STA	1,FORCE		
00526'000534		JMP	COMPM	AROUND WE GO AGAIN	
00520 100534	. THE EO			A PRESSURE SEG. ALREADY E	XISTS
005071000000	THE FOL	Ø	Incons Ir	A PRESSURE SECT REPERT E	
00527'000000	EX3:	A REAL PROPERTY AND A REAL			
00530 000021-	PRADD:	.PRES		ALLET WEAD	
00531'030021-	EXIST:	LDA		JLIST HEAD	
00532'151112		MOVL#	2,2,SEC	INO SECMENTS	
00533'001400		JMP	0.3	INO SEGMENTS	
00534'054773		STA	3, EX3		
00535'024454		LDA	1,NB		
00536'020452		LDA	ØINP		
00537'101300		MOVS	0,0	Shine and the state of the stat	
00540'107000		ADD	0,1	INPNB	
00541 '034767		LDA	3, PRADD	PREVIOUS HEAD IN AC3	
00542'021000	ANCHOR:	LDA	0,0,2	JIST WORD	

00543'106414		SUB#	0,1,52R	SAME NPNB?
00544 000403		JMP	CHAIN	INO: KEEP GOING
00545 010762		ISE	EX3	
00546'002761		JMP	eEX3	JGOOD EXIT
00547'155400	CHAIN:	INC	2,3	
00550'175400		INC	3,3	
00551 '031002		LDA	Contraction of the second second	INEN SEG.
00552'151112		MOVL#	2,2,520	
00553'002754		JMP	eEX3	;END OF CHAIN; EXIT!
00554'000766		JMP	ANCHOR	
00554 000100	THE FOL	The second se		A NEW PRESSURE SEG. ENTRY
00555'020466	NEWEN:	LDA	ØFORCE	
00556'101005	INC. N.C.IN.	MOV	0.0.SNR	
00557 000724		JMP	AGAIN	
00560'030020-		LDA		TRY EMPTY P. LIST
		and the second se		STAT ENTITY - LIST
00561'151112		MOVL#		MUST USE VIRGIN MEMORY
00562'000407		JMP		
00563'021002		LDA	and the second se	JOLD LINK
00564'040020-		STA	FRI (1990) 11 (1991) 1971 (1990)	REVISE EMPT POINTER
00565'034021-		LDA		CURRENT HEAD OF P. LIST
00566'055002		STA	3,2,2	INEW LINK
00567 050021-		STA	and the second se	JINSERT NEW P. SEG.
00570'000430	-	JMP	REST	JNOW PUT IN DATA
00571 0300245	FRMEM:	LDA		INEXT FREE LOCATION
00572'0200255		LDA	Ø. MEM	SHIGHEST MEMORY
00573'024452		LDA	A REAL PROPERTY AND A REAL	SWORDS NEEDED
00574'147000		ADD	2,1	
00575'122513		SUBL#	1,0,5NC	;OVERFLOW?
00576'000416		JMP	ALLOK	;NO
000012		.RDX	10	
00577'0060045		JSR	e.MESS	JPUT OUT MESSAGE
00600'001406'		MOVFL		
00601 000310		200		
00602'000574		380		
000010		.RDX	8	
00603 000700		JMP	AGAIN	
00604*000000	CHAR1:	Ø		
00605.000000	XP:	0		
00606'000000	YP:	Ø		
00607 '000000	AC28:	0		
00610'0000000	NP:	Ø		
00611 000000	NB:	0		
00615.000000	ZIMM:	0		
00613'000000	C5:	0		
00614'0440245	ALLOK:	STA	1. M7	REVISE FREE POINTER
00615'020021-		LDA	Ø, .PRES	
00616'041092		STA	0,2,2	
00617 050021-		STA	2. PRES	
00620'020423	REST:	LDA		INORMAL FORCE
00621 '041001	REST1:	STA	0,1,2	
00622 020422		LDA	Ø, MOMNT	MOMENT

00623'041003 00624'024765 00625'020763 00626'101300 00627'123000 00630'041000 00631'030756 00632'021000 00633'100000

STA LDA LDA MOVS ADD STA LDA LDA COM 0,3,2 1,NB 0,NP 0,0 1,0 0,0,2 2,AC2B 0,0,2 0,0

INPNB IHEAD OF GROUP IBLOCK POINTER ICONTROL WORD

00634'034412		LDA	3, PFLAG		
00635'163400		AND	3,0		
		COM	0,0		
00636'100000					
00637 '041000		STA	0,0,2		SSURE FLAG
00640'0060325		JSR	e.REB2	REBOX;	UPDATE FX, FY
00641 000642		JMP	AGAIN		
00642'000015	CRR:	15			
00643 000000	FORCE:	0			
00644'000000	MOMNT:	0			
00645 000006	SIZPR:	6			
00646'177377	PFLAG:	177377			
00647 000000	XA:	Ø			
09650'009000	XB:	0			
00651 '000000	YA: YB:	0			
00652 000000	LNG:	0			
00654 000000	XD:	0			
00655'000000	YD:	0			
00656'000000	xcc:	ø			
00657 000000	YCC:	ø			
00660'000000	HI:	ø			
00661 '000000	LO:	Ø			
	1				
00662 030725	COMPM:	LDA	2,AC28		
00663 024725		LDA	1.NP		
00664 0060305		JSR	e.PON1		
00665'040762		STA	Ø,XA		
00666'044763		STA	1.YA		
00667 024721		LDA	1.NP		
00670'0060275		JSR	e.LENG		
00671 040762		STA	Ø,LNG		
00672'021000		LDA	0,0,2		
00673'0340265		LDA	3. MSKR		
00674'163400		AND	3,0		
00675'125400		INC	1+1		
00676'122415		SUB#	1,0, SNR		THEFT LA STREET
00677'126400		SUB	1.1	IMUST BE	FIRST CORNER
and have been been and the base of the second		and the second sec	and the second sec		

00700'0060315

00701 034745

00702'162400

00703'034746

00704'166400

00705 040747

00706'044747 00707 021001

00710 024675

00711'122400

00712'040744

00713'021003

LDA SUB STA JSR YD YCC STA STA JSR XD XCC

JSR

LDA

SUB

LDA

SUB

STA

STA

LDA

LDA SUB

STA

LDA

1, YP 1.0 Ø,YCC JSIGNED MULTIPLY SMUL

3XB-XA

JYB-YA

MID-POINT

SXC

SYC

Ø,HI 1,10 SMUL

e.PON2

3, XA

3,0

3,1

3, YA

Ø,XD

1,YD

1,XP

1,0

0,1,2

Ø,XCC

0,3,2

		1.0.4	0.117	
00727 030731		LDA	2,HI	
00730 034731		LDA	3,L0	ADD O DD MUMDEDS
00731'167022		ADDZ	and the second second	JADD 2 DP NUMBERS
00732 151400		INC	2,2	
00733'143000		ADD	2.0	
00734'176400		SUB	3,3	ALE CATALIES
00735'101113		MOVL#	0,0,SNC	; NEGATIVE?
00736'000405		JMP	NONEG	JNO
00737 124405		NEG	1,1,SNR	
00740'109401		NEG	0,0,SKP	
00741'100000		COM	0,0	
00742'176520		SUBZL	3,3	
00743'030710	NONEG:	LDA	2,LNG	
00744'073101		DIV		
00745'030676		LDA	2.FORCE	
00746'102400		SUB	0.0	
00747 073301		MUL		
00750 175005		MOV	3,3, SNR	
00751'000404		JMP	BIT8	
00752'124405		NEG	1,1, SNR	
00753'100401		NEG	0,0,SKP	
00754'100000		COM	0.0	
00755'0300265	BIT8:	LDA	2. MSKR	TAKE MIDDLE 8 BITS
00756'143700		ANDS	2,0	
00757 125300		MOVS	121	
00760'147400		AND	2,1	
00761 107000		ADD	0.1	RESULT IN ACL
00762'044662		STA	1.MOMNT	202020200
00763'002417		JMP	etwt	
00764'000000	SMUL3:	Ø		
00765 054777	SMUL:	STA	3. SMUL3	
00766'027400	51106.	LDA	1,00,3	THE PARTY AND AND ADDRESS OF A DREAM OF A DR
00767 033401		LDA	2,01,3	
00770'176400		SUB	3,3	
00771 125112		MOVL#	1,1,SZC	
00772'157000		ADD	2,3	
00773'151112		MOVL#	2,2,SZC	
00774'137000		ADD	1,3	
00775'102400		SUB	0,0	
00776'073301		MUL		
00777'162400		SUB	3,0	
01000'034764		LDA	3. SMUL3	
01001'001402		JMP	2,3	
01002'000471'	TWT:	TWIT		
	1			
	3 APPLI	ED LOAD	INPUT (N	·UM-)
	1000	CTA	0.01.007	
01003'050437	LODE:	STA	2.BLKPT	
01004'0060045		JSR	e.MESS	
01005'001431'		NEWX		

01007'000113 01010'006003- XLOD: 01011'006004-01012'0060045 01013'001445' 01014'000416 01015'000113 01016'000772 01017'006005-

01006'000175

JSR JSR JSR SMES 270. 75. JMP JSR

XLOD

e.NGAT

125.

e.SIGN ;GET SIGN OF LOAD e.BRNG ;GET LOAD e.MESS

010001000000		1.0.4	-			
01020'030422		LDA	2.BLKPT		1211 11 2000	
01021'045023		STA	1,23,2	; PUT IT	IN LIST	
	3	100				
01022'0060045		JSR	e.MESS			
01023'001437'		NEWY				
01024'000175		125.				
01025'000067	110 22	55.				
01026'006003-	YLOD:	JSR	e.SIGN			
01027 006004-		JSR	₽.BRNG			
01030'0060045		JSR	e.MESS			
01031'001445'		SMES				
01032 000416		270.				
01033'000067		55.				
01034'000772		JMP	YLOD			
01035'006005-		JSR	e.NGAT			
01036'030404		LDA	2, BLKPT			
01037 045024		STA	1,24,2			
01040'002401		JMP	econt			
01041 177777	CONT:	CONTR				
	;					
01042'000000	BLKPT:	Ø				
	;					
	;					
	J DISP	LACEMENT	CONTROL I	ROUTINE		
	1					
01043 0060045	MOVE:	JSR	0.MESS			
01044'001577'		BMES				
01045'000144		100.				
01046'000144		100.				
01047 0060105		JSR	e.CURS	J SELECT	BLOCK	
01050'001154'		CHRC	ercuns	J JELEU	DECON	
01051 001155		XDM				
01052'001156'		YDM	A LITC			
01053'0060175		JSR	e.HITC			
01054'001155'		XDM				
01055'001156'		YDM	HOUT	TOY AC		
01056'000765		JMP	MOVE	ITRY AGA		
01057 020475		LDA	Ø,CHRC	; IS IT		
01060 034473		LDA		JIF SO H		
01061 116415		SUB#		: UNHOON	C DCM	
01062 000531		JMP	FNSH			
01063 050014-		STA		BLOCK B		
01064'044013-		STA		JAND NUM		
01065'176520		SUBEL	3,3	JGEN A		
01066'054012-		STA	3. MFLG	3 ALERT	DCM	
	3					
	3	ACCEPT D	ISPLACEMEN	NTS		
	;					
01067 0060035		JSR	e.PAGE			
01070'0060045		JSR	e.MESS			
		and a state of				

01071'001457' 01072'177470 01073'000764 01074'0060045 01075'001477' 01076'000341 01077'000733 01100'0060045 01101'001515' 01102'000226 DMS1 -200. 500. JSR DMS2 225. 475. JSR DMS3 150.

e.MESS

e.MESS

-						
	01103'000620		400.			
	01104'006003-	CGX:	JSR	P.SIGN		
	01105'006004-		JSR	.BRNG		
	01106'0060045		JSR	0.MESS		
	01107'001445'		SMES	C THEOD		
	01110'000764		500.			
	01111'000620		400.			
			JMP	CGX		
	01112'000772			and the second s		
	01113'006005-		JSR	e.NGAT		
	01114'044007-		STA	1XCGD		
	a sources a	1				
	01115'006004\$		JSR	e.MESS		
	01116'001531'		DMS4			
	01117'000226		150.			
	01120'000536		350.			
	01121'006003-	CGY:	JSR	e.SIGN		
	01122'006004-		JSR	e.BRNG		
	01123'0060045		JSR	e.MESS		
	01124'001445'		SMES			
	01125'000764		500.			
	01126'000536		350.			
	01127'000772		JMP	CGY		
	01130'006005-		JSR	e.NGAT		
	01131'044010-		STA	1. YCGD		
		3				
	01132'0060045		JSR	e.MESS		
	01133'001614'		DMS7			
	01134'000226		150.			
	01135'000454		300.			
	01136'020451		LDA	Ø, PLUS		
	01137'006004-		JSR	e.BRNG		
	000005		.BLK	5	INEED 5 SPACES	TO USE .BRNG
	01145'044011-		STA	1, SYCL		
	0					
	01146'0060045		JSR	e.MESS		
	01147'001545'		DMS5	CTILLES		
	01150'000310		200.			
	01151 000372		250.			
	01152'002667		JMP	@ CONT		
	01132 002001		Juir	e cont		
	01153'000305	ESKP:	"E+200		JADD PARITY BIT	Contraction of the second
		CHRC:	and the second s		THE THAT IT OIL	. CINDIT, 19215
	01154'000000	Contraction of the second s	0			
	01155'000000	XDM:	0			
	01156'000000	YDM:	Ø			
		;				
		5		2 00014		
	01157'054432	SGN:	STA	3.GOBK		
	01160'006015\$		JSR	e.GETT	J + OR - FIRST	
	01161'040431		STA	Ø,SIGN		
	01162'024425		LDA	1,PLUS		
	011001000000		COLUMN AL	CALC:	A LATE POPT	

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01164'000406 01165'024423 01166'106415 01167'000403 01170'034421 01171'001401 01172'034417 01172'034417

01163'106415

JMP LDA SUB# JMP LDA JMP LDA JMP

OK1:

SUB#

OK1 ; OUT IF + 1,MNUS Ø,1,SNR ;MUST BE -OK1 ; OUT IF -3,GOBK

1,3

0,3

3,GOBK

0,1, SNR : MUST BE +

- 110-1

3 3 -3 01174'054415 BRNG: STA 3,GOBK 01175'020415 . LDA Ø, SIGN 01176'0060125 JSR E.PRN2 PRINT SIGN 01177 0060145 JSR 8.DBIN ; X LOAD IS IN ACI 3,GOBK LDA 01200'034411 01201 001405 JMP 5,3 ; ---01202'020410 NGAT: LDA Ø,SIGN SIGN OF NEW LOAD ;ASCII -01203 030405 LDA 2, MNUS SUB# 0,2, SNR 01204'112415 01205'124400 NEG 1,1 01206'001400 JMP 0,3 3 **+ 01207 000053 PLUS: ** --MNUS: 01210'000055 01211 000000 GOBK: Ø 01212'000000 SIGN: Ø 3 FNSH: SUB 01213'126400 1 > 1 1, MFLG ; TURN OFF FLAG 01214'044012-STA 01215 0060045 JSR e.MESS 01216'001562' DMS6 01217 177324 -300. 01220'001130 600. 01221 002620 JMP **e**CONT 3 TEXT1: .TXT *SU 01222 052523 01223'043122 RF 01224'041501 AC 01225'020105 E 01226'051120 PR OP 01227'050117 01230'051105 ER TI 01231 044524 ES 01232 051505 01233'000000 * TEXT2: • TXT *TY 01234'054524 01235 042520 PE 01236 000000 * • TXT *FR 01237 051106 TEXT3: 01240'041511 IC 01241 044524 TI ON 01242'047117 01243'000000 * .TXT *DE 01244 042504 TEXT4: 01245 040506 FA 01246'046125 UL

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01247'020124 T 01250'052050 (T 01251'050131 YP 01252'020105 E 01253'020043 # 01254'024460 0) 01255'000000 * 01256'051120 TEX

Ø) * TEXT5: .TXT

*PR

01257 050117	OP				
01260'051105	ER				
01261 054524	TY				
01262'021440	#				
01263'003040	*				
01264'047111	TEXT8:	.TXT	*IN		
01265'052520	PU				
01266'020124	Т				
01267 043117	OF				
01270'042040	D				
01271'051511	IS				
01272'040524	TA				
01273'041516	NC				
01274 020105	E				
01275'047101	AN				
01276'020104	D				
01277'047506	FO				
01300'041522	RC				
01301'020105	E				
01302'047125	UN				
01303'052111	IT				
01304'000123	S*				
01305'040503	TEXT9:	.TXT	*CA		
01306'052125	UT				
01307'047511	IO				
01310'035116	N:				
01311 000000	*				
01312'047117	TEX10:	.TXT	*ON		
01313'054514	LY				
01314'047040	N				
01315'046525	UM				
01316'042502	BE				
01317 051522	RS				
01320'043040	F				
01321 047522	RO				
01322'020115	M				
01323'020061	1				
01324'044124	TH				
01325'047522	RO				
01326'043525	UG				
01327'020110	н				
01330'030065	50				
01331'030060	00				
01332'040440	A				
01333'046114	LL				
01334'053517	OW				
01335'042105	ED				
01336'000000	*				
01337'044127	TEX11:	• TXT	*WH		
01340'052101	AT	A FR			
012011000000	D				

01343'047531 01344'020125 01345'040527 01346'052116 01347'052040 01350'044510 01351'020123 01352'042514

01341'042040

01342'020117

YO U WA NT Т HI S LE

D

0



						C 1C2
-						C-163
	01353'043516	NG				
	01354'044124	ТН				
	01355'052040	T				
	01356'020117	0				
	01357'042522	RE				
	01360'051120	PR				
	01361'051505	ES				
	01362'047105	EN				
	01363'037524	Τ?				
	01364'000040	*				
	01365'044127	TEX12:	•TXT	* W H		
	01366'052101	AT				
	01367 044440	I				
	01370'020123	S				
	01371'044124	TH				
	01372'020105	E				
	01373 047125	UN				
	01374'052111	IT				
	01375 053440	W				
	01376'044505	EI				
	01377'044107	GH				
	01400'020124	T				
	01401 043117	OF				
	01402'051040	R				
	01403'041517	00				
	01404'037513	K?				
	01405'000040	*				
	01406'046407	MOVFL:	•T×T	*<7>M		
	01407 046505	EM				
	01410'051117	OR				
	01411'020131	Y				
	01412'053117	ov				
	01413 051105	ER				
	01414'046106	FL				
	01415'053517	OW				
	01416'000000	*				
	01417 050007	TOBIG:	•TXT	*<7>P		
	01420'042522	RE				
	01421'051523	SS				
	01422'051125	UR				
	01423'020105	E				
	01424'047524	TO				
	01425'020117	0				
	01426'040514	LA				
	01427'043522	RG				
	01430'000105	E*	TYT	AND		
	01431'042516	NEWX:	.TXT	*NE		
	01432'020127	W				
	01433'020130	X				
	01434'047514	LO				
	01435'042101	AD				
	01436'000040	*	TYT	THE		
	01437 042516	NEWY:	•TXT	*NE		
	01440'020127	W				

01440'020127 W 01441'020131 Y 01442'047514 L0 01443'042101 AI 01444'000040 M 01445'051440 SM 01446'043511 I0

Y LO AD * SMES: IG

* S

.TXT

01447 020116	N			
01450'044506	FI			
01451'051522	RS			
01452 020124	T			
01453'046120	PL			
01454'040505	EA			
01455'042523	SE			
01456'000040	*			
01457'047111	DMS1:	.TXT	*IN	
01460'052520	PU			
01461 '020124	T			
01462'044506	FI			
01462 044500	XE			
	D			
01464'020104	A STATE OF A			
01465'046102	BL			
01466'041517	oc			
01467'020113	K			
01470'044504	DI			
01471'050123	SP			
01472'040514	LA			
01473'042503	CE			
01474'042515	ME			
01475'052116	NT			
01476'000123	S*			
01477'031050	DMS2:	•TXT	*(2	
01500'054105	EX			
01501'030520	P1			
01502'020066	6			
01503'051511	IS			
01504'047440	0			
01505'042516	NE			
01506'051440	S			
01507 051103	CR			
01510'042505	EE			
01511'020116	N			
01512'047125	UN			
01513'052111	IT			
01514'000051)*			
01515'020130	DMS3:	• TXT	*X	
01516'042503	CE			
01517'052116	NT			
01520'047522	RO			
01521'042111	ID			
01522'042040	D			
01523'051511	IS			
01524'046120	PL			
01525'041501	AC			
01526'046505	EM			
01527 047105	EN			
01530'000124	T*			
01531'020131	DMS4:	.TXT	*Y	
01532'042503	CE			
01533'052116	NT			
01000 000110				

01234 041222	RU
01535'042111	ID
01536'042040	D
01537'051511	IS
01540'046120	PL
01541'041501	AC
01542'046505	EM

PO

A15341047500



01543 047105 EN 01544'000124 T * DMS5: 01545'044506 01546'044516 NI 01547'044123 SH 01550'042105 ED 01551'053454 28 AI 01552'044501 01553'044524 TI 01554 043516 NG 01555'040440 A 01556'020124 T 01557 047503 CO 01560 052116 NT 01561 000122 R* 01562'047125 DMS6: 01563'047510 HO 01564 945517 OK 01565'042105 ED 01566'042040 D 01567 046503 CM 01570'026440 -01571 020055 -AT 01572'052101 01573'041440 C 01574'047117 ON 01575'051124 TR 01576'000000 * BMES: 01577 042523 LE 01600'042514 01601 052103 CT 01602 041040 B LO 01603'047514 01604'045503 CK 01605 044054 +H 01606'052111 IT 01607 040440 A 01610'054516 NY 01611 045440 K 01612'054505 EY 01613'000000 . DMS7: 01614'041440 YC 01615'041531 LE 01616'042514 01617'020123 S 01620'042502 BE 01621'053524 TW 01622 042505 EE 01623'020116 N 01624'047515 MO

01625 042526

01626'020123

VES

.TXT *F1 • TXT *UN *SE • TXT • TXT * C

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01627'000000 * .END

; FROUTINE TO EXTERNALLY MOVE A FIXED BLOCK ; .ENT . DCM .DISB, .MESS, .REBX, .PFLG .EXTD .MOI, .FORD, .ALLB, .XCGD, .YCGD .EXID .SYCL, .MFLG, .SIEP, .DMBN, .DMBP .EXTD . ZREL 00000-000002. .DCM: MOVE .NREL ; 00000,000000 RET3: 0 00001 '000001 1 DMCT: ; 00002 054776 MOVE: STA 3, RET3 I. MFLG ; CHECK IF DCM 00003 0240135 LDA 00004'125005 MOV 1,1,SAK E KEI3 JGO BACK NO DCM 00005'002773 JMP JONLY EVERY . SYCL CY 00006 014773 DSZ DMCT 00007 002771 e RET3 ; GO BACK NOT RIGHT JMP 00010 0340125 3. SYCL LDA 00011 054770 STA 3, DMCT JRESET COUNTER 1, XCGD JAPPLIED X DISP 00012 0240105 LDA 00013'135000 MOV 1,3 1,1,SEC ;CHECK FOR SIGN 00014'125112 MOVL# 00015'124400 NEG 1 . 1 2. . DM8P 00016'0300165 DCMX: LDA 00017 '021002 :XC(LOW) LDA 0,2,2 00020 175112 MOVL# 3,3,520 FLIT JMP JWAS NEGATIVE 00021 000405 1,0, SNC 00022 123023 ADDE 00023 000417 JMP OK 00024 011001 ISZ 1,2 ; INCREMENT XC(HIGH) 00025 000405 JMP CHECK 00026 124400 FLIT: NEG 1,1 00027'123022 ADDZ 1,0,SZC 00030 000412 JMP OK JDECREMENT XC(HIGH) DSZ 1,2 00031 015001 CHECK: STA 1,20,2 JDEL XC 00032 045020 STA 0,2,2 00033'041002 LDA 1. DMBN 00034 0240155 JRE-CLASSIFY THIS BLOCK JSR e.REBX 00035 0060035 3, .PFLG 00036'0340045 LDA MOV 3,3, SNR 00037'175005 00040 0060015 JSR e.DISB JMP NUT 00041 000403 STA 1,20,2 JDEL XC 00042'045920 OK: STA 0,2,2 INEW XC(LOW) 00043 041002 3 1. YCGD JAPPLIED Y DISP

00044'0240115 NUT: LDA 1,.YCG 00045'135000 MOV 1,3 00046'125112 MOVL# 1,1,SE

00047'124400 00050'0300165 DCMY: 00051'021004 00052'175112 00053'000405 00054'123023 00055'000417 NEG LDA LDA MOVL# JMP ADDZ JMP 1,1,SEC ;AS ABOVE 1,1 2,.DMBP 0,4,2 ;YC(LOW) 3,3,SEC FLITS 1,0,SNC 0KS

-					
	00056 011003		152	3,2	INCREMENT YC(HIGH)
	00057 000405		JMP	CHECS	
	00260'124400	FLITS:	NEG	1,1	
	00061 123022		ADDE	1,0,520	
	00062 0000412		JMP	OKS	
	00063 015003		DSZ	3,2	DECREMENT YC(HIGH)
	00064 045021	CHECS:	STA	1,21,2	
	00065 041004		STA	8,4,2	
	00066 0240155		LDA	1 DMBN	
	00067 0060035		JSR	e.REBX	JRE-CLASSIFY
	00070 0340045		LDA	3. PFLG	
	00071 175005		MOV	3,3, SNR	
	00072 0060015		JSR	e.D158	FLOT JUST THIS BLOCK
	00073 000403		JMP	CLIT	
	00074'045021	OKS:	STA	1,21,2	;DELYC
	00075 041004		STA	8,4,2	INEW YC(LOW)
		1			
	00076'060477	CLIT:	READS	Ø	ICHECK FOR SW Ø
	00077'101122		MOVEL	0,0,SZC	JOFF = MESS
	00100'000405		JMP DUI	DE	
	00101 '0060025		JSR	P.MESS	
	00102'000117'		MOMS		
	00103'000144		100.		
	00104'000144		100.		
	00105'0060055	DUDE:	JSR	e.MOT	
	00106'0060065		JSR	e.FORD	
	00107 0060145		JSR	e.STEP	
	00110'0300165		LDA	2. DMBP	JGET BLOCK POINTER
	001111102400		SUB	0,0	SET ALL TO Ø
	00112'041020		STA	0,20,5	JDEL X
	00113'041021		STA	0,21,2	JDEL Y
	00114'041022		STA	0,22,2	JDEL AL
	00115'0060075		JSR	e.ALLB	JUPDATE CONTACTS
	00116'002662		JMP	erets	JGO BACK
		1			
	00117'047515	MOMS:	.TXT	*M0	
	00120'042526	VE			
	00121 020104	D			
	00122'000041	1*			
		1			

.END

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Voegele, Michael D

Rational design of tunnel supports: an interactive graphics based analysis of the support requirements of excavations in jointed rock masses / by Michael D. Voegele, Department of Civil and Mineral Engineering, University of Minnesota, Minneapolis, Minn. Vicksburg, Miss. : U. S. Waterways Experiment Station ; Springfield, Va. : available from National Technical Information Service, 1979.

v, [516] p. ill.; 27 cm. (Technical report - U. S. Army Engineer Waterways Experiment Station; GL-79-15)

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References: p. R-1 - R-9.

 Excavation. 2. Interactive graphics. 3. Jointed rock.
 Rock masses. 4. Tunnel supports. I. Minnesota. University. Dept. of Civil and Mineral Engineering. II. United States.
 Army. Corps of Engineers. III. Series: United States. Waterways Experiment Station, Vicksburg, Miss. Technical report ; GL-79-15. TA7.W34 no.GL-79-15