

UNITED STATES DEPARTMENT OF THE INTERIOR

GEOLOGICAL SURVEY

Pacific Enewetak Atoll Crater Exploration (PEACE) Program Enewetak Atoll, Republic of the Marshall Islands

Part 4: Analysis of borehole gravity surveys and other geologic and bathymetric studies in vicinity of OAK and KOA craters

edited by

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U.S. Geological Survey

Open File Report 87-665

Prepared in cooperation with the Defense Nuclear Agency And funded

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PACIFIC ENEWETAK ATOLL CRATER EXPLORATION (PEACE) PROGRAM ENEWETAK ATOLL, REPUBLIC OF THE MARSHALL ISLANDS

Part 4: Analysis of borehole gravity survey and other geologic and bathymetric studies in vicinity of OAK and KOA craters

CHAPTER 1:

INTRODUCTION TO PART 4 OPEN-FILE REPORT

By

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GENERAL REMARKS

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The Pacific Enewetak Atoll Crater Exploration (PEACE) Program was established and funded by the Defense Nuclear Agency (DNA) to resolve a number of questions for the Department of Defense (DOD) about the geologic and material-properties parameters of two craters (KOA and OAK), formed by nearsurface bursts of high-yield thermonuclear devices on the northern margin of Enewetak Atoll, (fig. 1-1), Marshall Islands, in 1958. The multidisciplinary studies conducted by the USGS in collaboration with the DNA, the Department of Energy (DOE), and other organizations during 1984 through 1987 were part of a much larger research initiative by the DNA to better understand the dynamic properties of strategic-scale nuclear bursts and the relevance of the Pacific Proving Grounds (PPG) craters to issues of strategic basing and targeting of nuclear weapons.

The reader is referred to the reports cited in the succeeding section for a detailed explanation of the events leading up to the PEACE Program and the collaborative roles of the USGS, other Federal agencies, and scientists and engineers from universities and private research laboratories.

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FIGURE 1-1. -- Map of Enewetak Atoll, Republic of the Marshall Islands (RMI), showing native names of principal islands and other features (military site name in parenthesis), and location of OAK, KOA, and MIKE craters. Inset map shows location of Enewetak within Pacific Ocean. Map from Henry, Wardlaw, and others (1986, p. 12, fig. 3).

PEACE PROGRAM REPORTS

This volume is the fifth and final volume of a series of four U.S. Geological Survey (USGS) Open-File Reports (Henry, Wardlaw, and others, 1986; Cronin, Brouwers, and others, 1986; and Henry and Wardlaw, 1986a) and one USGS Bulletin (Folger, 1986a) documenting geologic and geophysical data, analyses, and interpretations for the PEACE Program. Syntheses for the geologic and material-properties models for the craters are found in Wardlaw and Henry (1986, Ch. 14) and Wardlaw (1987, Ch. 7, this Report). The materialproperties studies themselves, which provide quantitative parameters for computational modeling, for the most part were not conducted by USGS personnel and are published elsewhere for the DNA (e.g., Blouin and Timian, 1986a, 1986b; Borschel, Klauber, and Earley, 1986; McClelland Engineers, 1986: Mueller, 1987; Patti and Schatz, 1987 [1988], in preparation, Schatz, Patti, and Melzer, 1987 [1988], in preparation, and Simons and others, 1984).

DATA ACQUISITION AND BASES

The PEACE Program was truly a multidisciplinary endeavor. Field work for the program on Enewetak Atoll was done in two parts, the Marine Phase (mid- to late summer, 1984) and the Drilling Phase (late winter through mid-summer, 1985). The primary and derivative PEACE Program data bases and framework groups consist of the elements shown in Table 1-1. For general discussion of the fieldwork and data-acquisition procedures for the Marine Phase, the reader is referred to Folger (1986b), and, for the Drilling Phase, to Henry, Wardlaw, and others (1986, p. 29-97). For more detailed information about the field and laboratory procedures employed for a specific data set, refer to the individual Chapters or volumes (see tbl. 1-1). Many of the derivative data sets and framework groups from the Drilling Phase utilized samples from the 32 deep and intermediate boreholes drilled from the M/V Knut Constructor in the Enewetak lagoon. These boreholes (figs. 1-2 and 1-3) provide a data base upon which the subsurface geologic framework is grounded and upon which interpretations made from the geophysical and material-properties studies must be validated.

A wide array of pre-PEACE Program data from the PPG was re-examined, including (but not limited to) the following:

- (1). Published accounts in USGS Professional Paper 260 series (see Emery, Tracey, and Ladd, 1954) from the initial geologic, geophysical, and oceanographic investigations in the Marshall Islands associated with the early phases of nuclear testing.
- (2). Published reports and raw data from the geologic and geophysical studies of the PACE, EXPOE, and EASI Projects, sponsored by the DNA and conducted on Enewetak by the Air Force Weapons Laboratory (AFWL) (Couch, Fetzer, and others, 1975; Henny, Mercer, and Zbur, 1974; Ristvet, Tremba and others, 1978; Tremba, Jones, and Henny, 1981; Tremba, Couch, and Ristvet, 1982; and Tremba, 1987). For example, some of the multichannel-seismic lines lines from EASI were reprocessed by Grow, Lee, and others (1986), and selected PACE/EXPOE boreholes were redescribed and analyzed stratigraphically and isotopically before the Drilling Phase actually got underway (Henry, Wardlaw, and others, 1986;

TABLE 1-1. -- Matrix of data bases and analyses from PEACE Program. In heading, CH = Chapter; under heading PHASE, Marine or Drilling connotes which phase the samples were obtained originally. The pilot gravity survey in the old borehole on Medren (ELMER) Island was conducted in April 1984, hence the asterisk (*) in the appropriate column. The geologic and material-properties models for the craters are presented in Wardlaw and Henry (1986, Ch. 14) and in Wardlaw (1987, Ch. 8, this Report). U.S. Geological Survey Open-File Report 87-665 is the current volume.

DATA GROOP	PRASE	PUBLICATION	CH1	REFERENCE CELATION
Bathymetric		Bult. 1678	•	Falsur Hommon and atheres (1986).
Мары	Marine	OF-87- 665	ŝ	Peterson and Benny (1987*).
			-	
Side-Scan Sonar	Marine	Buil. 1678	8	Folger, Rubb, and others (1986).
lwagery			-	
Single-Channel			_	
Seismic Reflection	Marine	Bull. 16/8	с	Robb, Foster, and others (1986).
M.1.4				
Selemic Reflection	Harle	8ull. 1678	D	Grow, Lee, and others (1986).
Seismic-Refraction	Martne	8uil. 1678	E	Ackermann, Grow, and Williams (1986).
Submarathla	Marlau	No.11 14.78		
Observations	Buth	08-86-555		Halley, Slater, And Others (1986); Slater Wouldy and others (1986);
			• •	marer, koudy, and others (1000).
	Marine	Bull. 1678	C	Halley, Major, and others (1986);
Debris/Ejecta	Harine	08-86- 555	3	Luduig, Halley, and others (1986);
	oricing	0F-8/- 665*	4	Polanskey and Ahrens (1987*).
Scuba Observations	Marine	Bull. 1678	н	Shilma, Kindiager, and others (1986).
Botrom Samples	Matula	OF-86- 555	10	Wardlaw, Benry, and Mariln (1986).
Boreholes	Drilling	OF-86 414	-	Henry, Wardliw, and othern (1986).
Lithostratigraphic				
Framework	Drilling	OF-86- 555	2	Wardlaw and Henry (1986a).
kiontration-li.				
Framework & Mixing	Drilling	OF-86-159		Cronin, Brouwers, and others (1986);
Studies	Difficing	UF-87- 665*	3	Broowers, Cronin, and Gibson (1986);
			•	
Geophysical Logs	Drtlling	0F-86- 555	7	Melzer (1986).
		0F-87- 665*	6	Truito (1987*).
Selsmic Reference				
Survey	Drilling	OF-80- 555	9	Tremba and Ristver (1980b).
Borehole Gravimetry	(rel) land	0F-86- 555	8	Beyer, Kistver, and Obernie-Lehn (1986
,	ortring	0F-87- 665#	2	Buyer (1987+),
			U	fruito (1987*).
Si-leatope	Martne	Bull. 1678	G	Halley, Ludwig and others (1986)
Pramework	Drilling	08-86- 535	3	Ladwig, Halley, and Simmone (1986).
X-Ray Mineralogy	Orilling	0F-86→ 555		• • • • • • • • •
			•	Tremba and Klatver (1980).
organic Geochemistry	Drilling	OF-86- 555	5	Risivet and Tremba (1986).
Insoluble Residues	Dr.E.D.Com	0 P 94 . 517		
	orrig	06.00- 333	6	Ristvet and Tremba (1986),
Radiation Chemistry	Urilling	OF-86- 555	12	Ristvet and Trambs (1946)
Electron-Sele	1			and the and treating (1760),
Кенопаласе	Mainly	0F-87- 6uj#	4	Polanskey and Abrow (1987#)
Crater Area	Drilling	OF-86- 555	4	fromba nul kternus (1063)
nottom Sampies	Sectoring	OF-86-555	10	Wardlaw, Henry and Marsia (1986).
			-	Patti and Schatz (1987) [1988?].
Crator Succession	Marine	Bull. 1678	A	Folger (19866).
oracer synthesis	Both	08-86- 555	14	Wardlaw and Henry (19866)
	Both	UF-87- 665+	7	Handlan (10070)

* OF-87-665 is the current report.

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FIGURE 1-2. -- Map of KOA crater area showing borehole sites (depicted by letters) and general bathymetric contours (contour interval in feet). Map modified from Henry and Wardlaw (1986b, fig. 1-2).



FIGURE 1-3. -- Map of OAK crater area showing borehole sites (depicted by letters) and general bathymetric contours (contour interval in feet). Map modified from Henry and Wardlaw (1986b, fig. 1-3).

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Cronin, Brouwers, and others, 1986; Halley, Major, Ludwig, and others, 1986).

- (3). A broad spectrum of unpublished archival material from the PPG made freely available to us by the DNA and the DOE. These data include preand post-shot survey maps of the OAK and KOA crater areas, both black-andwhite and color, stereographic aerial photographs, other kinds of aerial photographs, and pictures made (both pre- and post-testing) from groundlevel of various crater features and man-made structures. The pre- and post-shot Holmes and Narver maps of OAK crater were digitized and form an essential part of the volumetric studies for the PEACE Program (Peterson and Henny, 1987, Chapter 5 of the current Report).
- (4). Other published reports, too numerous to cite here.

SYNOPSIS OF CHAPTERS OF CURRENT VOLUME

This Open-File Report consists of seven Chapters. The interrelationship of each Chapter to the overall data base is depicted in Table 1-1. Salient points of each Chapter are summarized below.

Borehole Gravity (Ch. 2; Beyer)

The borehole gravity measurements from the southwest transect of OAK crater and in the Medren (ELMER) Island borehole provide a critical set of data for bulk density and porosity of both the undisturbed stratigraphic sequence and the sediments and rock that were affected by the OAK event.

Significant densification, porosity dimunition, and mass removal are indicated for discrete intervals within the boreholes in the central-crater region of OAK. Zones in which these phenomena are indicated correspond closely to the geologic crater zones and provide strong corroborative evidence for their integrity.

One of the primary goals of the gravimetry was to determine whether densification of the shallow substrate in the crater-flank region (or "wings") could account for the measured lowering of the sea floor. This is particularly critical because the bulk of the volume of the apparent crater lies within its flank region. Gravity analysis conducted in the upper parts of transition-zone boreholes (OQT-19 and ORT-20, see fig. 1-3) demonstrates conclusively that the materials (sediment, rock, and rock debris) are only slightly denser than comparable intervals of materials in reference boreholes OOR-17 and OSR-21. In fact, only perhaps about 15 percent of the documented lowering of the sea floor in the crater wings region can be explained by densification alone. Thus, for the wings region, processes other than densification clearly are also involved.

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Paleontologic Evidence for Mixing (Ch. 3; Cronin, Gibson)

Paleontologic analysis of the upper 1,200 ft of strata on Enewetak established the division of the upper part of the carbonate cap into twelve discrete biostratigraphic zones (named zones AA, the youngest, through LL, the oldest -- see also Cronin, Brouwers, and others, 1986, and Brouwers, Cronin, and Gibson, 1986). For the current study, additional samples from stratigraphically undisturbed boreholes from the OAK crater area (i.e., from the reference boreholes) were examined to refine the local zonation and to more closely resolve key biostratigraphic boundaries. Several of these boundaries, combined with physical stratigraphic datums, form surfaces or marker horizons that are Lagrangian (see Chapter 6), permitting employment of a powerful tool in the analysis of crater evolution.

The microfossil studies of OAK reference-borehole plus crater samples provides significant new information about the timing and methods of emplacement¹ of materials from various biostratigraphic zones within the materials that particlly infill the crater itself. This includes for the first time identification of sediments that were either at or within a few centimeters of the pre-event lagoon floor. These new data have furnished quantatative estimates of material from each zone (or group of zones) admixed in the crater fill. These estimates include volumes and percentages of materials originating from the deeper stratigraphic zones not involved in the excavation of the initial crater itself and from shallower geologic units as well. The editors emphasize that materials from stratigraphically shallower zones pose a real problem of differentiation. For example, how does one separate material that may have been emplaced from, say, zone CC from material within CC that has not moved? Therefore, estimated volumes or percentages of material that may have been piped or otherwise moved from these shallow zones muy be underestimated. perhaps grossly.

Electron Paramagnetic Resonanace Studies (Ch. 4; Polanskey, Ahrens)

EPR spectrometry was applied to measure the peak-shock stress to which calcitic materials were subjected during the OAK event. Most of the samples analyzed can be characterized as either unshocked or very heavily shocked, with few samples showing intermediate states. Samples of the "transition sand" from OPZ-18 show the greatest concentration of very highly shocked material, interpreted as originating in the proximity of ground zero and plastered onto the walls of the excavational crater. Because of subsequent collapse of the excavational crater walls and dilution by mixing with lessshocked or unshocked materials, this lining, as a discrete stratigraphic unit, is identifiable only in the OPZ-18 borehole. Suprisingly, none of the 26 samples from the ground-zero borehole OBZ-4 showed significant shock damage. However, a zone containing less concentrated, very highly shocked material can be recognized in the three transition-zone boreholes studied (OCT-5, OET-7, and OFT-8), and its base occurs at progressively shallower depths away from ground zero.

¹ The term emplacement is used as a generic term to describe the deposition of material transported from one point to another without reference to the mechanism involved.

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Bathymetric Studies of OAK Crater (Ch. 5; Peterson, Henny)

Three pertinent base maps were digitized and processed with a computer to facilitate analysis of the changes in the sea-floor bottom topography (bathymetry) and corresponding volumes in the area affected by the detonation of the OAK device (June 29, 1958) and by subsequent, longer term geologic processes. These maps are: (1) the pre-shot Holmes and Narver (H&N) survey, completed three days prior to the burst; (2) the post-shot H&N map, surveyed 47 to 67 days after the burst; and (3) the USGS map, made during the Marine Phase of the PEACE Program, 26 years after the burst.

The USGS map, in the format presented by Folger, Hampson, and others (1986), was not amenable for comparison with the two H&N maps (even undigitized) primarily because the USGS depth contours are given in meters rather than feet. The irregular area common to all three base maps is shown in Figure 5-1.

OAK is a strongly asymmetric crater; part of the asymmetry is a geologic function of the reef being on one side of surface ground zero and the atoll lagoon on the other. Many independent lines of evidence demonstrate that the excavational crater was appreciably smaller and more nearly circular than the current (or apparent) crater. Using the standardized digitized data for the common area of the base maps, three pairs² of vertical-difference contour maps were prepared. These maps show that: (1) the pre-shot topographic (geologic) features significantly influenced not only the evolution and final size/shape of the crater but also the initial distribution and subsequent reworking of debris from the OAK event; (2) the area of greatest downward displacement of the sea floor between the two post-shot base maps is that of the inner crater; and (3) the entire map area was lowered (and not uniformly) an average of 23 ft by 67 days after the burst and by another 12 ft during the next 26 years. As the surface of the lagoon and crater floor in this area was lowered, areas of positive-difference in relief (i.e., those areas that were higher post-shot than pre-shot) also decreased from about 27 percent by 67 days to about 14 percent 26 years later.

Two notes of caution must be clearly understood in using these maps for quantitative estimates for cratering calculations. The first is that there is no Lagrangian marker for the pre-event lagoon floor. The second is that the debris volumes estimated from these maps are understated simply because the apparent crater of OAK extends beyond the areas mapped, including the USGS map, which ecompasses the largest area.

Following the glossary presented in Henry and Wardlaw (1986a), the apparent crater is defined as the locus of the zero-difference contour line surrounding a crater -- viz, the locus of points where the effects of an explosion can no longer be detected when the pre-event contours are compared with the post-event contours (fide, B.L. Ristvet.)

² A negative- and a positive-relief-difference (called Δ -relief) isopachous map was constructed for each combination of two base maps.

Constraints on Densification and Piping for OAK (Ch. 6; Trulio)

As mentioned previously, it is established from a wide array of data that the excavational crater of OAK had an appreciably smaller radius than that of the apparent crater. Because crater volume is a radius-squared function, it is evident that most of the volume of OAK is contained within its flank or "wing" area. What is (are) the significant mechanism(s) reponsible for forming the wings of the large apparent craters in the PPG? Trulio presents a number of different models dealing with the PEACE Program data bases and makes a number of inferences about these mechanisms based on these models.

Using the data base from the gravimetry (Chapter 2), Trulio applies mathematical analyses to the data, from a purely physical viewpoint, and verifies Beyer's conclusion that densification (or, in Trulio's terminology, "simple subsidence") accounts for just a small part of the formation of the wings of OAK crater. As a best estimate, only about 8 percent of the seafloor drop on the wings can be attributed to density increases caused by the burst.

Another explanation for part of the observed sea-floor lowering phenomenon is piping, or movement (driven by gravity and density differences) of a sediment/water slurry through conduits (cracks, fissures, etc.) to generally shallower depths or to the surface through vents to form "sand volcanoes". That piping occurred associated with the OAK and KOA bursts, particularly in the central crater region, is supported by independent lines of evidence (see Chapter 7 for discussion). However, at issue are: (1) the role of piping relative to other mechanisms to account for the drop in the sea floor; and (2) the amount of material transported by this mechanism. Mean values for the density of material piped up to the sea floor from beneath OAK can be derived from the combination of sea-floor base maps and gravimetry profiles. If correct, this model poses limitations on the amount of material transported out of the crater by piping. The best estimate based on this model is that the piped and residual materials differed by only about 0.2 g/cc, a density difference that can drive piping, in Trulio's words, "but weakly". Trulio cautions that the sequence of events leading to the transport of piped material out of the crater is subject to interference at many points.1

It is suggested that plastic flow also should be considered as a plausible mechanism to account for most of the phenomenon of sea-floor lowering. Trulio points out, however, that little is known about the displacement field around a flow crater.

¹ See caveat in italics on page 1-8. The editors also emphasize that the observed "subsidence" or sea-floor lowering on the wings of the Enewetak craters studied is not reasonably attributable to one mechanism operating alone. The lowering was caused in part by densification, in part by piping (certainly upwardly and probably laterally as well), probably in part to plastic flow, and possibly to other mechanisms that may not have even been thought of yet.

Additional Studies of Geologic Crater Models (Ch. 7, Wardlaw)

The final Chapter provides an integration of the new information from the various studies presented in the current Open-File Report with the previously developed analyses of PEACE Program data. Of particular interest to the material-properties community is the formulation of a set of material-properties units for the normal stratigraphic (geologic) sequence and a discussion of the relationship of these units to the sedimentary packages presented in Wardiaw and Henry (1986a, 1986b).

Using available evidence, the pre-event geology beneath the OAK and KOA crater areas is reconstructed, including paleotopographic contour maps of several of the more significant subsurface datums. Wardlaw points out that topographic differences of the pre-event Holocene ground surfaces (i.e., the pre-1958 lagoon, reef, and island surface) between the KOA and OAK area produced differences in the surface configurations of the two craters. Differences in cementation and structural competency of key stratigraphic intervals beneath the surface ground zeros of KOA and OAK and the effects of these differences on the development of the two craters are summarized.

A study of the thinning of the stratigraphic units influenced by OAK and KOA is presented. A more comprehensive interpretation of the models for these two craters given in Wardlaw and Henry (1986b) is developed based primarily on the inferred pre-shot elevation of certain datums and thicknesses of stratigraphic intervals in contrast to their post-shot attributes. The case is ade that movement of material laterally ("lateral flow") may account for much of the "subsidence" and formation of the wings.

An idealized succession of cratering and depositional events is presented.

ACKNOWLEDGEMENTS

We, the editors, extend a special note f appreciation to the following people, without whom this program would have not been possible. Lt. Col. Robert F. Couch, Jr. (U.S. Air Force and DNA Program Manager for the PEACE Program) and Byron L. Ristvet and Edward L. Tremba (both of S-Cubed Division of Maxwell Laboratories), and Robert W. Henny (Air Force Weapons Laboratory) were full scientific collaborators with us during the PEACE Program. All four of these geologists logged extensive on-site experience in the PPG prior to the current program and were principal investigators in all of various phases of the earlier AFWL investigations on Enewetak. And, in a real sense, they represent a vital component of the record of cratering studies on Enewetak. Their expertise and geotechnical knowledge were invaluable to the current program, and we owe them a profound debt of gratitude. Couch, Ristvet, and Tremba served (alternatively) as Chief Scientists aboard the <u>Knut Constructor</u> during the Drilling Phase of the program, and all three were <u>on-site</u> during parts of the earlier Marine Phase.

We would like to thank also the authors of the Chapters of the current Open-File Report for their timely response to our needs in editing, compiling, and finalizing this volume and for their input for synthesizing the diverse data bases.

We are indebted also to John F. Schatz and L. Stephen Melzer of SAIC for constructive exchange of scientific and technical information for this volume. Melzer was on-site with us as Chief Scientist during part of the Drilling Phase field work, and it was his observations along with that of the geologists and paleontologists studying the OAK ground-zero borehole aboard the drill ship that demonstrated the reality of piping of materials from zones from far below the excavational crater into the sediment forming part of the crater fill.

The plates for Chapter 2 were laid out by James MacCornack, S-Cubed, Albuquerque, and printed by the Defense Nuclear Agency Printing Plant, Kirtland AFB, New Mexico. We thank Leonard MacDonald, head printer, for his assistance.

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CHAPTER 2:

ANALYSIS OF BOREHOLE GRAVITY SURVEYS AT OAK CRATER

by

L. A. Beyerl

INTRODUCTION

Borehole gravity (BHG) surveys were made in selected PEACE Program boreholes at OAK crater on Enewetak Atoll because they provide the only means to directly and accurately measure in situ bulk density of large volumes of rock and sediment that surround the boreholes and to provide data to calculate the total porosity of these materials2. The differences between the density and porosity of undisturbed atoll materials and the sediment and rock involved in the excavational and apparent craters are crucial to understanding various cratering phenomena. In addition, accurate and representative density and porosity measurements of undisturbed atoll materials are important for nuclear-event calculations. The nature of BHG measurements, rationale for siting BHG boreholes, field techniques, and preliminary (apparent) BHG density data and calculated porosity values are given in Beyer, Ristvet, and Oberste-Lehn (1986).

This report presents the models used to correct the apparent (BHG) density and porosity data for large-scale lateral density changes across the reef margin (due to natural facies changes) and for smaller-scale lateral density changes due to cratering phenomena. Corrected BHG density data and calculated porosity values are described in terms of their modification due to cratering processes.

Ancillary topics include: (1) general results of the BHG survey in the E-1 borehole on Medren (ELMER) Island (Appendix 2-1), (2) brief comparison of estimates of density and porosity from BHG, gamma-gamma, and neutron logs, and (3) relationship between grain density and BHG porosity in undisturbed atoll materials. A short description of how average interval grain density was determined from the x-ray mineralogy and organic analyses studies of core samples is found in Appendix 2-2.

The locations of OAK crater and E-1 and F-1 deep boreholes referred to later in this chapter are shown in Figure 2-1. Locations of boreholes drilled at OAK crater during the PEACE Program are given in Figure 2-2, along with a table that summarizes pertinent information about the boreholes in which BHG surveys were made. Locations of two cross sections presented later in the chapter also are shown in Figure 2-2.

lBranch of Sedimentary Processes, U.S. Geological Survey, Menlo Park, CA.

2Bulk density and total porosity are abbreviated as density and porosity in this Chapter. Porosity is calculated from a combination of <u>in situ</u> density and grain-density data derived from x-ray mineralogic analyses.

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FIGURE 2-1. - Map of Enewetak Atoll showing locations of OAK, KOA, and MIKE craters and Medren (ELMER) Island. Deep boreholes E-1 and F-1 drilled in 1951 and 1952 by the USGS and AEC (Ladd and others, 1953; Ladd and Schlanger, 1960) and referred to in this paper are shown by "X"'s on the inset maps.

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The efficacy of BHG surveys to determine subtle density differences at OAK crater depends in great part on the precision of the field measurements. This is determined by making repeated Wg/Wz measurements over the same depth interval. Of 98 intervals surveyed in six boreholes, 8 percent were repeated four or more times, 81 percent were repeated three times, 10 percent were repeated two times, and 1 percent were not repeated due to operational constraints. These repeated Wg/Wz measurements indicate that the precision of the surveys is quite high and fully adequate for the purposes of the OAK study (fig. 2-3). Standard deviations of repeated measurements are given in column 3 of Tables 2-2 through 2-7 (located at the end of the current Chapter), are illustrated graphically on BHG density and porosity plofiles in subsequent figures, and are explained in Appendix 8-2 of Beyer, Ristvet, and Oberste-Lehn (1986).

BOREHOLE GRAVITY ANALYSIS

The analysis of BHG measurements at OAK crater follows the only logical path available in the absence of independent data such as a detailed surface gravity anomaly map and reliable density data from gamma-gamma and/or core measurements. BHG measurements are corrected for recognizable lateral density variations so that the corrected BHG densities are reasonably accurate measures of the atoll materials within a few tens to a few hundreds of feet of each surveyed borehole. Then, comparisons of density (and porosity) can be made between different boreholes in and near OAK crater.

Corrections can be made rationally for submarine topography (Beyer, Ristvet, and Oberste-Lehn, 1986), for large-scale lateral density charges across the reef margin that are caused by natural facies changes, and for smaller-scale lateral density changes related to cratering processes. A summary of the range of corrections calculated and applied to the BHG surveys is given in Table 2-1. Individual corrections are presented in Tables 2-2 through 2-7, located at the end of the Chapter1.

Corrections cannot be made for even smaller-scale lateral density changes on the order of tens to about a hundred feet distant from each borehole, because data needed to model these very small density changes were beyond the scope of the PEACE Program. We will note where these very small-scale effects may be present. Neglect of them does not impair the objectives of the BHG phase of the PEACE Program.

Please note that these corrections are computed as vertical gravity gradients which, when multiplied by 0.25 k, where k is the Neutonian gravitational constant, become density corrections in g/cm3. Lateral density variations that cause a downward positive vertical gravity gradient result in a positive density correction, whereas a downward negative gradient causes a negative density correction (see Appendix 8-2 of Beyer, Ristvet, and Oberste-Lehn, 1986).

1A11 tables are located at the end of the Chapter.

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(expressed in g/cm^3)

FIGURE 2-3. - Distribution of 93 sets of repeated $\Delta g/\Delta z$ measurements during borehole gravity surveys in OAK boreholes, expressed in g/cm³. Mean is .009 g/cm³ and is a measure of the high quality of the borehole gravity surveys (Beyer, 1968; Black and Herring, 1983).

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LARGE-SCALE LATERAL DENSITY CHANGES ACROSS REEF MARGIN

A substantial body of work by many investigators at modern Pacific atolls has shown that forereef and reef core facies generally are more highly cemented (and therefore denser) than lagoon facies, and that atoll reefs generally prograde seaward (e.g., Buigues, 1985). These relationships are believed to be present along the northwest margin of Enewetak Atoll according to B. L. Ristvet, who provided the author with a sketch of the probable distribution of facies and densities across the reef margin at OAK crater. Other PEACE Program studies (Folger, 1986a) and earlier work at Enewetak, especially deep boreholes E-1 and F-1 and the XEN series of boreholes on Engebi Island (Ladd and others, 1953; Ladd and Schlanger, 1960; Couch and others, 1975), led to this assessment of atoll margin structure. Densities provided by Ristvet were modified slightly using the BHG densities from the E-1 borehole on Medren Island (see Appendix 2-1).

Deeper density contrasts (e.g., between the volcanic core and overlying carbonate rocks of the atoll) and possible incomplete isostatic compensation of the stoll also can affect the vertical gravity gradients (and BNG densities). Corrections for these possible effects are almost certainly negligibly small and, if determined, would cause only a very small, constant dc-type shift of all density data. The absence of even a rudimentary surface gravity anomaly map and more detailed deep borehole data prevent any attempt to examine these effects.

The two-dimensional density model prepared for the atoll margin at OAK crater is shown in Figure 2-4. Vertical gravity gradient corrections were calculated for the two-dimensional model with a well-established algorithm (Talwani and others, 1959) that has been modified for borehole gravity applications. These corrections are given in column 5 of Tables 2-2 to 2-7 and probably are unnecessary but their magnitudes needed to be evaluated.

CORRECTION FOR LATERAL DENSITY CHANGES DUE TO CRATERING PROCESSES

Lateral density variations due to cratering processes also can affect the BHG densities and, therefore, were evaluated. The model used to correct for these crater-related lateral density changes was developed along the southwest transect from OPZ-18 to OOR-17 by using BHG densities (corrected for submarine topography and large-scale density changes across the atoll margin) and a correlation cross section prepared by D. Oberste-Lehn and modified by B. R. Wardlaw (fig. 2-5; correlation cross section CD, fig. 2-6, also was prepared by Oberste-Lehn and Wardlaw). The density model is shown in Figure 2-7 and was assumed to have circular symmetry about OPZ-18. Trial gravity calculations taking into account the departure of OAK crater from circular symmetry about OPZ-18 (based only on correlation cross sections) showed that the assumption of circular symmetry is valid. The size of the corrections due to crater-related lateral density changes is so small that the question of true three-dimensionality versus circular symmetry about OPZ-18 is academic. The question of the actual crater density structure along cross section CD remains. A very careful sea floor gravity survey or more BHG drillholes and surveys would shed light on this question.





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FIGURE 2-5. - Cross section AB extends from south-southwest to north-northeast through markers in OBZ-4 extending to the right and markers in OPZ-18 extending to the lef zone boundaries H/I and J/K, and crater geologic zones α_1 through Y (see Wardlaw a by D. Oberste-Lehn and modified by B. R. Wardlaw. BHG density profiles are superi scale labels "1.90" correspond to the positions of the surveyed boreholes in the c



ast through OAK crater (fig. 2-2). Section is broken at center with geologic to the left. Correlation lines tie disconformities 1 through 6, biostratgraphic Wardlaw and Henry, 1986a,b). This section and that of Figure 2-6 were prepared are superimposed on the section for the six surveyed boreholes. The density is in the cross section. Vertical exaggeration from ORT-20 to ODT-6 is 2X.







est-southeast through OAK crater (fig. 2-2). Data are identical to those described for this section. Boreholes OJT-12, ONT-16, and OMT-15 are projected into section as shown

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In order to perform density calculations using the model shown in Figure 2-7, the assigned densities must be recast as density contrasts relative to the surrounding medium. The density of the surrounding atoll material was determined from the BHG density profiles of reference boreholes OOR-17 and OSR-21 which were assumed to be unaltered by crater-related processes. The density model for the surrounding medium is summarized in Table 2-8, and values are shown along the left side of Figure 2-7. These reference density values were subtracted from laterally juxtaposed crater density model elements to arrive at a density contrast model.

Corrections to the BHG densities in OPZ-18, OTG-23, OQT-19, and ORT-20 were calculated from the density contrast model using a well-established algorithm for three-dimensional density elements (Plouff, 1976) modified for borehole gravity application. These corrections proved to be very small (column 6, tables 2-2 through 2-7), which could be predicted from the gentle dips of the density element boundaries as shown in Figure 2-7.

CORRECTED BOREHOLE GRAVITY DENSITY AND POROSITY AND COMPARISON WITH GAMMA-GAMMA AND NEUTRON DATA

Tabular and graphical summaries of BHG density and porosity with error estimates, grain densities with error estimates and interval-averaged density and porosity from gamma-gamma and neutron logs are presented in Tables 2-2 through 2-7 and Figures 2-8 through 2-13. Open-hole well log curves also are shown in Figures 2-8 through 2-12. Interval-averaged neutron porosity is not graphically displayed in Figures 2-8 through 2-12 because of a systematic error that has made all values too large. Interval grain density profiles are derived from individual grain density values, examination of open-hole well logs, and descriptions of cores and samples, sedimentary packages, and boreholes (Henry and Wardlaw, 1986; Wardlaw and Henry, 1986a; Holloway and Young, 1986). Errors in interval grain density are only estimates that may be increased or decreased with further information about the mineralogy of individual intervals, particularly intervals with both aragonite and calcite.

A number of questions about the reliability of the gamma-gamma density and neutron porosity logs run in OAK boreholes were raised during the analysis of the borehole data. Corrected BHG density and porosity provide a reliable standard against which gamma-gamma and neutron logs can be evaluated. In OAK boreholes, the differences between BHG density and interval-averaged gammagamma density decrease with increasing depth below the sea floor (fig. 2-14). This result agrees with the well-documented body of literature from the petroleum industry, which indicates that shallow-penetration radiation well logs, such as the gamma-gamma and neutron logs run in OAK boreholes, perform poorly in loosely consolidated, highly permeable sediments where formation damage caused by rotary drilling is almost always substantial.

Relatively good correspondence between BHG and gamma-gamma data is obtained for intervals deeper than 600 ft below sea level in OOR-17 and OQT-19, and deeper than 500 ft below sea level in ORT-20 where drill-induced borehole and formation damage is less because sediments are somewhat more consolidated than at shallower depths (figs. 2-8, 2-10, and 2-11). More specifically, Figure 2-14 suggests that gamma-gamma density departs



DEPTH BELOW SEA FLOOR (FEET)

Asterisks indicate density and porosity are plotted where values are outside error regions of BHG density and porosity. logs from left to right are gamma-ray, multi-channel sonic, intervals where gamma-gamma log not available due to drillpipe. Interval averages of gamma-gamma disconformities and facies boundaries, sedimentary packages, and BRG and gamma-gamma porosity for FIGURE 2-8. - Open-hole well logs, BHG and gamma-gamma density, core and interval grain density, neutron porosity, caliper, gamma-gamma density, and gamma-gamma density correction. Well borehole OOR-17 (left to right).





packages, and BHG and gamma-gamma porosity for borehole OPZ-18 (left to right). Well logs from left to right are gamma-ray, neutron porosity, caliper, gamma-gamma density, and gamma-gamma density correction. Asterisks indicate intervals Interval averages of gamma-gamma density an FIGURE 2-9. - Open-hole well logs, BHG and gamma-gamma density, core and interval grain density, porosity are plotted where values are outside error regions of BHG density and porosity. where gamma-gamma log not available due to casing. sedimentary disconformities and facies boundaries,

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Interval averages of gammaborehole OQT-19 (left to right). Well logs from left to right are gamma ray, multi-channel sonic, disconformities and facies boundaries, sedimentary packages, and BHG and gamma-gamma porosity for gamma density and porosity are plotted where values are outside error regions of BHG density and neutron porosity, caliper, gamma-gamma density, and gamma-gamma density correction. Asterisks FIGURE 2-10. - Open-hole well logs, BHG and gamma-gamma density, core and interval grain density, indicate intervals where gamma-gamma log not available due to casing.

porosity.





borehole ORT-20 (left to right). Well logs from left to right are gamma ray, caliper, gamma-gamma density, and gamma-gamma density correction (partial). Asterisks indicate intervals where gammagamma log not available due to casing. Interval averages of gamma-gamma density and porosity are disconformities and facies boundaries, sedimentary packages, and BHG and gamma-gamma porosity for FIGURE 2-11. - Open-hole well logs, BHG and gamma-gamma density, core and interval grain density, plotted where values are outside error regions of BHG density and porosity.

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for borehole OSR-21 (left to right). Well logs from left to right are gamma ray, caliper, and FIGURE 2-12. - Open-hole well logs, BHG and gamma-gamma density, core and interval grain density, disconformities and facies boundaries, sedimentary packages, and BHG and gamma-gamma porosity gamma-gamma density. Interval averages of gamma-gamma density are plotted where values are outside error regions of BHG density and porosity.

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FIGURE 2-14. - Differences between BHG density and interval averages of gammagamma density versus depth below sea floor (bsf). Seventy differences are from boreholes OOR-17, OPZ-18, OQT-19, ORT-20 and OSR-21.

unsystematically from BHG density by as much as ± 10 percent at depths shallower than about 400 ft below the sea floor. At depths greater than about 500 ft below sea level, gamma-gamma density appears to vary from BHG density by about 2 percent or less.

Cross plots between BHG porosity and neutron porosity, gamma-gamma porosity and neutron porosity, and gamma-gamma porosity versus BHG porosity confirm that the neutron data are not adequate (fig. 2-15). These cross plots, along with Figure 2-14, emphasize the need for caution in the use of the gamma-gamma logs for quantitative density or porosity evaluation. Furthermore, bulk density and total porosity data derived from core measurements should be viewed skeptically if they differ significantly and systematically from corresponding BHG density and porosity.

NATURAL DENSITY AND POROSITY VARIATIONS OF ATOLL MATERIALS

Natural variations of density and porosity of atoll materials represent the background "noise" through which density and porosity changes caused by cratering phenomena must be determined. At the volume scale of core samples of several cubic feet, a broad range of values of densities and porosities from virtual sea-water-filled voids to dense crystalline carbonate is expected. At the volume scale of BHG studies (hundreds of thousands of cubic feet--an appropriate scale for studies of large craters), one expects the range of values of natural densities and porosities to narrow considerably because of the averaging effect. This is confirmed by the BHG surveys at OAK crater where the range of densities in reference boreholes OOR-17 and OSR-21 is not great.

At shallow depths in OOR-17, OSR-21, ORT-20 and OQT-19, density fluctuations are substantial but can be averaged to give nearly identical values (fig. 2-16). Based on the similarity of averaged BHG densities for OOR-17, OSR-21, ORT-20, and parts of OQT-19, the "noise" problem connected with natural density and porosity variations is believed to be small. However, close correspondence between individual BHG intervals from borehole to borehole can not be expected because of natural variations of density in porosity.

A general systematic relationship exists between BHG porosity and interval grain density based on data from reference boreholes OOR-17 and OSR-21 (fig. 2-17). Back reef sediments dominated by aragonite have higher porosities than sediments dominated by calcite. Effects on porosity caused by mechanical compaction and grain-size distribution and uncertainty about values of interval grain density may account for some or all of the scatter of points in Figure 2-17. Nevertheless, the rate of change of porosity with respect to aragonite content, as estimated by the dashed line, is almost identical to that found by Schmoker and Hester (1986) in a study of the late Pleistocene Miami Limestone. However, porosity values of the Miami Limestone are lower than Enewetak back-reef sediments by about 15 percent for equivalent aragonite content, emphasizing the different geologic settings of these two locations.

If bulk density is held constant, the mineral volume increase accompanying simple transformation of aragonite to calcite causes porosity to decrease by about 5 percent (line A in fig. 2-17). It is clear from Figure





- Cross plot of gamma-gamma porosity and neutron porosity, each averaged over BHG depth intervals Cross plot of BHG porosity and gamma-gamma porosity averaged over equivalent depth intervals Data are from columns 10, 13 and 14, Tables 2-2 through 2-7. (A).
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FIGURE 2-16. - Averages of BHG density and porosity fo.



resity for selected large interval. Porosities are in parentheses.

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FIGURE 2-17. - Interval grain density versus BHG porosity. Data from reference boreholes OOR-17 and OSR-21. Line A shows decrease in porosity (due only to mineral volume increase) during transformation of an aragonitebearing sediment to calcite-bearing sediment, assuming 100% aragonite sediment has porosity of 58.5%, and bulk-density is held constant.

2-17 that calcite solution and recrystallization, deposition of externally derived carbonate cement, some compaction, and other processes during welldocumented cycles of atoll emergence and submergence (Ristvet and others, 1974; Tracey and Ladd, 1974) also have contributed to porosity loss in the back-reef sediments around OAK crater.

DENSITY, POROSITY, AND MASS CHANGES RELATED TO CRATERING PHENOMENA

A corrected BHG density and porosity model of the south-southwest transect of OAK crater is shown in Figure 2-18. This generalized model closely, but not exactly, follows the disconformity, facies changes, and geologic crater zone correlations defined by Wardlaw and Henry (1986a,b). The density elements of this model are based on the interval divisions of the BHG surveys that were selected during field work prior to knowledge of the exact downhole locations of the geologic boundaries. Intervals of BHG density and porosity have not been divided to correspond to the geologic boundaries because such divisions would be arbitrary in the absence of gravity station readings at the downhole locations of the geologic boundaries. Furthermore, BHG density and porosity are based on mass/volume characteristics that may or may not coincide with divisions based on the geologic characteristics of the sediments. This is clearly seen in Figure 2-5 where a significant number of major BHG density changes occur between, rather than at, the geologic boundaries defined by Wardlaw and Henry (1986a,b). Lack of exact depth correspondence of geologic and density/porosity models does not interfere with comparisons of geologically equivalent intervals between boreholes (figs. 2-16 and 2-19). (Application of the borehole gravity data to the geologic interpretation of OAK crater is expanded in Chapter 7 of the current Open-File report.)

A primary goal of the BHG phase of the PEACE Program was to determine if densification in crater-flank regions could account for observed sea-floor subsidence. BHG surveys were made in transition-zone boreholes OQT-19 and ORT-20 to investigate possible densification. There is considerable variation of BHG density and porosity in the upper parts of these boreholes but averages over larger intervals show that the sediments are not now appreciably denser than in the reference boreholes OOR-17 and OSR-21 (fig. 2-16).

Because documented subsidence of the sea floor at OQT-19 and ORT-20 cannot be explained by densification of the upper few hundred feet of underlying sediments alone, mass displacement from this region and densification of deeper materials probably occurred. Selective removal of finer fractions in this way could be investigated by comparing grain-size distributions of core samples from OQT-19 and the reference boreholes. Slight but definite densification and porosity decreases are present at greater depths in OQT-19 (figs. 2-16 and 2-19).

Unmistakable densification and porosity diminution are inferred in boreholes OTG-23 and beneath 292 ft below sea level in OPZ-18 (figs. 2-16 and 2-19). Independently documented mass transport (Wardlaw and Henry, 1986b) also can be quantified with the BHG density and porosity data. For example, the mass columns at OQT-19 and OPZ-18 are mass deficient by 3 to 5 percent and 6 to 8 percent, respectively, when compared to the mass column at reference



FIGURE 2-18 - BHG density and porosity model for south-southwe



south-southwest transect at OAK crater. Porosities are in parentheses.



FIGURE 2-19 - Averages of BHG density and porosity for sele



ity for selected large intervals. Porosities are in parentheses.

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boreholes OOR-17 and OSR-21. Density and porosity changes over the large intervals shown in Figures 2-16 and 2-19 are summarized in Table 2-9.

Many more types of comparisons are possible but are beyond the intent of this paper, which is to provide accurate representative density data and calculated porosity values for atoll materials at the six surveyed boreholes and to present a general density model of the crater.

SUMMARY

Borehole gravity surveys were conducted at OAK crater to obtain accurate large-volume estimates of <u>in situ</u> bulk-density and total porosity of atoll materials beneath and beyond the crater. Reliable density and porosity measurements of undisturbed atoll materials provide important geologic information about these young, loosely consolidated back-reef sediments, and predictions of pre-event material-property conditions for nuclear event calculations. Accurate measurements of differences between the density and porosity of undisturbed atoll materials and the sediment and rock involved in the excavational and apparent OAK craters are crucial to understanding the cratering phenomena at OAK and the mechanics of large crater formation.

Six boreholes were drilled and successfully logged with a borehole gravity meter along a 6,000-foot southwest transect from the bathymetric center of OAK crater (fig. 2-2). Gravity measurements were made in these cased boreholes, generally at spacings of 20 to 35 ft. To obtain reference values for the density and porosity of undisturbed reef-forming material for comparison with material disturbed by cratering processes, gravity surveys were conducted in boreholes OSR-21 and OOR-17, separated by 562 ft, and located approximately 5,500 and 6,050 ft south-southwest of the bathymetric center of OAK crater. Possible densification caused by suspected subsidence on the crater flank just outside the excavational crater was investigated by gravity surveys made in boreholes 00T-19 and 0RT-20. These boreholes were separated laterally by 404 ft, and located approximately 1,400 and 1,800 ft south-southwest of the bathymetric center of OAK crater. Gravity surveys were made in borehole OPZ-18 at the bathymetric center of OAK crater, and in OTG-23 located 759 ft south-southwest of OPZ-18 to measure densification beneath the excavational crater and the density of fill within the excavational crater.

The ability of BHG surveys to determine subtle density differences between reef materials located at different locations inside and outside OAK crater depends crucially on the precision of field measurements. Consequently, great effort was devoted to insure that requisite precisions were achieved (Beyer, Ristvet, and Oberste-Lehn, 1986). Repeated measurements show that the precision of BHG densities averages about .01 g/cm3, which is fully adequate for the purposes of the OAK study (fig. 2-3).

BHG measurements permit examination of large volumes of material surrounding the borehole, which means that larger-scale, more distant, lateral density changes are sensed, along with smaller-scale, local density changes that occur within tens to a few hundred feet of the borehole. To obtain BHG density and porosity of atoll materials immediately surrounding each borehole, corrections were calculated and applied for: (1) submarine topography out to a distance of 103 statute miles using bathymetric charts of various scales (Beyer, Ristvet, and Oberste-Lehn, 1986); (2) generalized two-dimensional density variations associated with large-scale facies changes and diagenesis across the reef margin (fig. 2-4); and (3) generalized three-dimensional density changes due to cratering processes (fig. 2-7). Corrections range from .067 to .156 g/cm3 for submarine topography, .018 to .025 g/cm3 for two-dimensional density variations across the atoll margin, and +.025 to -.021 g/cm3 for generalized three-dimensional density changes due to cratering (tables 2-1 through 2-7).

The following conclusions can be drawn from the corrected BHG densities (fig. 2-5), the derived crater density model (fig. 2-18), and comparisons of corrected BHG density and porosity of geologically equivalent intervals along the south-southwest transect from OOR-17 to OPZ-18 (figs. 2-16 and 2-19):

- 1. Large natural variations of density and porosity of atoll materials are well-known from numerous geological observations at scales of cubic inches to hundreds of cubic feet. Serious concern was expressed prior to this study that these natural variations of density and porosity would obscure those due to cratering phenomena. The shallow portions of the BHG density profiles from OOR-17, OSR-21, ORT-20, and OQT-19 suggest that this concern may be well-founded when attempting to compare shallow vertical intervals in different boreholes that are on the order of tens of feet thick (fig. 2-5). However, averages of BHG densities over larger vertical intervals show that natural variations of density and porosity tend to average out so that valid lateral comparisons of geologically equivalent intervals can be made (figs. 2-16 and 2-19). Also, over the depths surveyed in OAK boreholes, the range of natural variation of density and porosity appears to decrease slightly with depth, allowing valid comparisons of smaller vertical intervals with increasing depth (fig. 2-5). These results are based solely on BHG surveys made on a trend nearly parallel to the reef along which facies and density changes are believed to be minimal.
- 2. The shallow section beneath the crater-flank region penetrated by OQT-19 and ORT-20 is not appreciably denser than the equivalent section penetrated by the more distant reference boreholes OOR-17 and OSR-21 (fig. 2-16). Crater-flank subsidence in this area cannot be explained by densification of this shallow section, but probably involved mass removal and densification of a larger vertical interval. Slight densification is evident at greater depth in OQT-19 but cannot be confirmed at greater depth in ORT-20 (figs. 2-16).
- 3. Atoll material penetrated by OTG-23 within the excavational crater is significantly denser over the surveyed intervals than the geologically equivalent sections penetrated by reference boreholes OPR-17 and OSR-21 and crater flank boreholes ORT-20 and OQT-19 (figs. 2-16, 2-18, and 2-19). Porosity reduction also has occurred as a result of cratering processes.
- 4. At the bathymetric center of OAK crater, the section penetrated by OPZ-18 is dominated by cratering effects. Major discontinuities of BHG density and porosity occur midway through crater zones b_{1a} and β_3 . Beneath the second discontinuity midway through zone β_3 , and extending at least to the J/K biostratigraphic boundary, a large amount of densification and porosity reduction has resulted from cratering processes (figs. 2-16 and

2-19). In the lower part of this densified interval, the low BHG densities compared to the indicated thinning of geologic units means that major amounts of mass have been removed (fig. 2-19). Densification and/or mass removal appears to extend beneath the depth of the BHG survey in OPZ-18, and may be evaluated by careful study of the lower portions of the gamma-gamma density logs from OAM-1/OAR-2, OBZ-4, OCT-5, and OOR-17.

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5. Mass deficiencies of about 3 to 5 percent at OQT-19, and 6 to 8 percent at OPZ-18 are indicated from mass column calculations that utilize BHG densities at OOR-17, OQT-19, and OPZ-18.

Other conclusions of this study are as follows:

- 1. In reference boreholes OOR-17 and OSR-21, back-reef sediments dominated by aragonite have higher porosities than materials dominated by calcite (fig. 2-17). The mineral volume increase that accompanies the transformation of aragonite to calcite is not sufficient to explain the observed decrease in porosity. Not surprisingly, other diagenetic processes, such as calcite solution and recrystallization and deposition of externally derived carbonate cement, must have contributed to (or dominated) the observed natural decrease of porosity that accompanies the transformation of aragonite to calcite in these back-reef materials.
- 2. The BHG survey in the E-1 borehole on Medrin (ELMER) Island provided BHG densities of the shallow section that differed only slightly from those measured across Enewetak Atoll at OAK crater (figs. 2-1 and 2-20). EHG densities of the deeper section in the E-1 borehole were important to the construction of the two-dimensional density model of the reef argin (fig. 2-4). The BHG survey in the E-1 borehole also revealed a cyclical pattern of density and porosity that may be due to diagenesis caused by repeated periods of atoll emergence and submergence since middle Miocene (fig. 2-20).
- 3. BHG measurements permit examination of volumes of materials measured in hundreds of thousands of cubic feet. Unlike conventional, shallow penetration gamma-gamma and neutron logging methods, the large volume of material examined by the BHG method makes it immune to formation damage or borehole rugosity that commonly occurs when drilling through loosely consolidated, highly permeable strata or alternating soft and hard beds. Not unexpectedly, BHG density and porosity in the OAK study are about an order of magnitude more reliable than the next most reliable density or porosity logging method, the gamma-gamma density log, at depths less than about 400 ft below the sea floor in the five OAK boreholes where comparisons were possible (figs. 2-8 through 2-12, 2-14, and 2-15).

This first BHG study of the carbonate deposits of an atoll island and of the materials beneath a large nuclear crater affirms the unique ability of borehole gravimetry to evaluate the density and porosity of heterogeneous and/or loosely consolidated geologic formations.

ACKNOWLEDGEMENTS

I am especially indebted to several key people. D. Oberste-Lehn, of R & D Associates, one of several people who initially recognized the potential value of borehole gravity surveys to the PEACE Program, solicited my involvement and worked closely with me during the past several years. Her persistent questions, stimulating ideas, and material support have been invaluable. Crucial leadership and considerable effort during the difficult field surveys were very ably provided by B. L. Ristvet of S-Cubed, Inc. Early recognition of the value of BHG surveys and strong backing from within the cratering community came from J. G. Trulio of Applied Theory, Inc. I also benefited greatly from information provided at various stages by J. J. Daniels, J. A. Grow, R. B. Halley, J. C. Hampson, T. W. Henry, L. S. Melzer, B. L. Ristvet, E. L. Tremba, J. G. Trulio, and B. R. Wardlaw.

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	Range of Corrections Expressed in g/cm ³ Due to								
Borehole	Submarine Topography	Large-Scale Lateral Density Changes Across Reef Margin	Smaller-Scale Latera Density Changes Relat to Cratering Process						
00R-17	.156 to .144	.021 to .019	negligible						
OPZ-18	.118 to .067	.025 to .021	.025 to021						
ogr-19	.145 to .135	.024 to .020	.008 to004						
ORT-20	.140 to .130	.023 to .020	.005 to005						
OSR-21	.137 to .126	.020 to .018	negligible						
OTG-23	.122 to .108	.022 to .024	.019 to .014						

TABLE 2-1. -- Range of corrections for lateral density changes calculated from submarine topography and density models for six BHG surveys at OAK Crater.

TABLE 2-2.--(on adjacent page) Bulk density, porosity, and grain density obtained from borehole gravity, gamma-gamma, and neutron measurements in borehole OOR-17 and from analysis of cores taken from OOR-17. Gammagamma and neutron data averages over depth intervals of BHG survey. Grain densities were calculated by the procedure described in Appendix 2-1 using data from Tremba and Ristvet (1986) and Ristvet and Tremba (1986).

14	60.2 57.6	58.8	53.3	58.9 61.5	58.1	52.3	54.6	0	0		。	0	58.0	59.6	60.6	59.4
13	47.8 47.2	50.0	43.6	46.1 51.9	1.01	46.8	49.1	.0	0			0	44.8	43.4	45.4	44.3
12	1.99 1.98	1.91	2.00	1.91	1.91	1.95	1.91	.0		0	.0		2.03	2.06	2.03	2.05
11	0.2	0.9	е. 0	2.0	6.0	0.5	1.3	0.2	0.1	0.6	0.5	6.0	0.2	0.5	0.2	0.4
10	50.2 49.3	51.6	41.5	54.0	46.8	47.2	45.8	43.7	43.0	41.0	44.5	45.9	46.8	44.7	46.0	46.3
6	.02	.02	.02	.02	.04	.04	.04	.02	. 02	.02	.02	.02	.02	.02	. 02	.02
8	2.87 2.83	2.79	2.75	2.86 2.86	2.76	2.76	2.76	2.71	2.71	2.72	2.73	2.78	2.84	2.85	2.86	2.86
7	1.947 1.943	1.882	2.036	1.901	1.951	1.944	1.967	1.976	1.987	2.027	1.974	1.976	1.993	2.036	2.018	2.013
10 -	 0 0	0			. 0	.0	0	.0	.0	.0	.0	.0	.0	.0	.0	.0
'n	.019 .019	.019	010 .	.020	.020	.020	.020	.020	.020	.020	.020	.021	.021	.021	.021	.021
4	.156	.154	.153	.152	.151	.150	.149	.149	.148	.147	.147	.146	.146	.145	.145	.144
m	.003 .008	.015	.006	.013 .013	.015	.009	.022	.003	.001	.010	.008	.016	.004	.009	.004	.007
2	1.772 1.769	1.709	1.864	1.700	1.780	1.774	1.798	1.807	1.819	1.860	1.807	1.809	1.826	1.870	1.852	1.848
-	212237. 237271.	271289.	289317.	317345. 345363.	363388.	388417.	417446.	446477.	477505.	505528.	528560.	560603.	603645.	645683.	683713.	713747.

2

4 Column Column Column Column

ŝ Column

Depth interval in feet below as level Apparent BHG denaity in g/cm³ (corrected for inatrument calibration and drift and earth tides). Apparent BHG denaity in g/cm³ (corrected for inatrument calibration and drift and earth tides). Vertical gravity gradient correction expressed in g/cm³ for submarine topography out to a radial diatance of 103.5 statute miles (166.7 Vertical gravity gradient correction expressed in g/cm³ for submarine topography out to a radial diatance of 103.5 statute miles (166.7 Vertical gravity gradient correction expressed in g/cm³ for submarine topography of to be two-dimensional) caused by geologic facies vertical gravity gradient correction expressed in g/cm³ for lateral density changes (assumed to be two-dimensional) caused by geologic facies vertical gravity gradient correction (expressed in g/cm³) for lateral density changes caused by crater-related processes and assumed to be vertical gravity gradient correction for submarine topography and lateral density variatione due to geologic facies end cratering waterical and to gravity gradient correction for submarine topography and lateral density variations due to geologic facies changes and cratering bBG density in g/cm³ after correction for submarine topography and lateral density variations due to geologic facies changes and cratering Column 6

processes. ~ "olumn

recomments for depth interval in g/cm³ (calculated from mineral and organic content percentages estimated from x-ray diffraction and loes-on-ignition analyses (see Appendix 2.2). Estamated uncertainry in mean grain density in g/cm³. BHG porceity in Percent calculated from BHG density/(column 7), grain density (column 8) and sea-water density uf 1.03 g/cm³. Uncertainry in BHG porosity in porosity percent (calculated from standard deviation of BHG density (column 4) and uncertainty in mean grain Column 8

σ Column

Column 10 Column 11

density (column 8).

Average gemme-gamme density for depth interval in g/cm³. Gamme-gamme porosity in percent calculated from gamma-gamme density (column 12), grain density (column 8) and sea-water density of 1.03 g/cm³. Average neutron porosity for depth interval in percent. 222 Column Column Column
TABLE 2-3.--(on adjacent page) Bulk density, porosity, and grain density obtained from borehole gravity, gamma-gamma, and neutron measurements in borehole OPZ-18 and from analysis of cores taken from OPZ-18. Gammagamma and neutron data averages over depth intervals of BHG survey. Grain densities were calculated by the procedure described in Appendix 2-1 using data from Tremba and Ristvet (1986) and Ristvet and Tremba (1986).

14			57.8 54.0	53.1 57.0	51.2	51.7 48.1	49.8 49.5	53.5	58.3	58.1	61.1	59.4
13	 		41.6 39.0	40.4 40.9	39.2	40.433.7	35.5	35.5	42.1	44.3	44.0	44.5
12	 		2.07	2.09 2.10	2.13	2.05 2.15	2.12	2.12	2.09	2.05	2.06	2.04
11	0.1	0.4 0.6	0.5 0.5	0.6 0.6	0.4	0.5	0.2	0.4	0.6	0.1	6.0	0.3
10	56.1 56.8 40.0	46.0 46.0	47.6 37.9	45.3 41.7	35.9	35.1 36.3	35. 4 36.3	35.8	44.4	44.2	49.9	45.0
6	.02	. 02	.02	.02	.02	. 02	.02	.02	.02	.02	.02	. 02
œ	2.81 2.83 2.83	2.83 2.82	2.81 2.80	2.81 2.84	2.84	2.74	2.72	2.72	2.86	2.86	2.87	2.85
7	1.812 1.807	1.956	1.962 2.130	2.003 2.086	2.191	2.139 2.107	2.122	2.115	2.048	2.052	1.951	2.031
Q	- 005 - 006	006	021 0.016	0.002 0.005	0.019	0.020 0.020	0.021	0.025	0.021	0.021	0.021	0.020
ſ	.021	.021	.022	.022	.023	.023	.024	.024	.024	.024	.024	.025
4	.067 .073	.080 .084	.090 .095	.098 .100	.103	.105	.109	.113	.114	.116	.117	.118
m	600. 100.	.010 .010	.008	.010	.007	600. 600.	.003	. 007	.011	.002	.017	.006
7	1.729 1.719	1.861 1.861 1.892	1.871 1.997	1.881 1.958	2.046	1.991 1.957	1.968 1.950	1.953	1.889	1.891	1.789	1.868
-	232262. 262292.	292322. 322352. 352382.	382457. 457492.	4 92522. 522552.	552582.	582610. 610637.	637672. 672 -707	707742.	742772.	772807.	807837.	837867.

Column 1 Column 2 Column 3 Column 4

Depth interval in feet below sea level Apparent BHG density in g/cm³ (corrected for instrument calibration and drift and earth tides). Apparent BHG density in g/cm³ (corrected for instrument calibration and drift and earth tides). Vertical gravity gradient correction expressed in g/cm³ for submarine topography out to a radial distance of 103.5 statute miles (166.7 Vertical gravity gradient correction expressed in g/cm³ for submarine topography out to a radial distance of 103.5 statute miles (166.7 Vertical gravity gradient correction expressed in g/cm³ for lateral density changes (assumed to be two-dimensional) caused by geologic facies Vertical gravity gradient correction expressed in g/cm³) for lateral density changes caused by crater-related processes and assumed to be vertical gravity gradient correction (expressed in g/cm³) for lateral density changes caused by crater-related processes and assumed to be vertical gravity in gradient correction for submarine topography and lateral density variations due to geologic facies changes and cratering winnerrical about Q22-18 (see Figure 2.7). Column 5

ø Column

Column 7

processes. Mean grain density for depth interval in g/cm³ (calculated from mineral and organic content percentages estimated from x-ray diffraction and Mean grain density and yeas (see Appendix 2.2). Estimated uncertainty in mean grain density in g/cm³. BHG porosity in percent calculated from BHG density (column 8) and sea-water density of 1.03 g/cm³. Uncertainty in BHC porosity in porosity percent (calculated from standard deviation of BHG density (column 4) and uncertainty in mean grain Column 8

Column 9 Column 10 Column 11

Coluan

Average gamma-gamma density for depth interval in g/cm³. Gamma-gamma porosity in percent calculated from gamma-gamma density (column 12), grain density (column 8) and sea-water density of 1.03 g/cm³. Average neutron porosity for depth interval in percent. 221 Column Column

TABLE 2-3.

TABLE 2-4.--(on adjacent page) Bulk density, porosity, and grain density obtained from borehole gravity, gamma-gamma, and neutron measurements in borehole OQT-19 and from analysis of cores taken from OQT-19. Gammagamma and neutron data averages over depth intervals of BHG survey. Grain densities were calculated by the procedure described in Appendix 2-1 using data from Tremba and Ristvet (1986) and Ristvet and Tremba (1986).

14				59.1	60.4	57.6	60.2	62.6	52.8	55.4	52.0	53.7	55.3	53.5	54.8	54.9	61.4	60.4	57.6	60.5	61.8	61.7
13			0	47.0	45.6	43.3	45.5	53.4	38.5	45.0	42.4	41.1	41.1	44.0	42.1	44.4	45.9	44.8	45.9	45.7	44.8	47.5
12				1.99	2.01	2.04	2.00	1.86	2.13	2.02	2.02	2.02	2.02	1.97	2.02	2.03	2.01	2.04	2.02	2.03	2.04	1.98
11	2.1	1.1	0.9	0.3	0.4	0.5	0.6	0.4	0.5	0.8	0.4	0.7	0.4	1.0	0.5	0.1	0.8	0.3	0.5	0.3	1.2	0.4
10	57.1	49.3	52.2	51.6	51.7	48.4	47.1	49.9	45.4	45.5	40.9	38.9	42.4	45.4	46.1	43.9	44.8	46.3	46.0	46.7	44.8	48.8
6	.02	. 02	.02	.02	.02	.02	.02	.02	.02	.02	.02	.02	.02	.02	.02	.02	.02	.02	.02	.02	. 02	.02
œ	2.84	2.83	2.83	2.84	2.83	2.81	2.81	2.81	2.82	2.83	2.75	2.71	2.71	2.71	2.74	2.83	2.84	2.86	2.86	2.87	2.86	2.84
7	1.807	1.942	1.890	1.906	1.900	1.948	1.972	1.922	2.008	2.011	2.046	2.056	1.998	1.948	1.951	2.040	2.030	2.013	2.018	2.011	2.041	1.956
9	004	0. UU4	0.	0.001	0.001	0.005	0.005	0.004	0.008	0.008	0.007	0.004	0.001	0.	100.0	0.001	001	002	002	002	002	002
ъ	.020	.021	.021	.021	.021	.021	.022	.022	.022	.022	.022	.022	.022	.023	.023	.023	.023	.023	.024	.024	.024	.024
4	.145	.134	.132	.131	.131	.131	.131	.132	.132	.132	.132	.133	.133	.133	.134	.134	.134	.134	.135	.135	.135	.135
m	.038	.020	.017	.006	.008	.009	.011	.008	.009	.014	.007	.012	.007	.017	.008	.002	.015	.005	600.	.005	.022	.008
7	1.646	1.787	1.737	1.753	1.747	1.791	1.814	1.764	1.846	1.849	1.885	1.897	1.842	1.792	1.793	1.882	1.874	1.858	1.861	1.854	1.884	1.799
-	138148.	1481/8. 178208.	208238.	238273.	273298.	298328.	328358.	358380.	380403.	403433.	433468.	468498.	498528.	528558.	558588.	588618.	618648.	648678.	678708.	708738.	738768.	768798.

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Depth interval in feet below sea level Column 1

Column 2 Column 3 Column 4

Column 5

Apporent WHG denarty in 9/20⁻¹⁰ (corrected for instrument calibration and drift and earth tides). Apporent WHG denarty in 9/20⁻¹⁰ (corrected for instruments expressed in 9/20⁻¹⁰. Standard deviation of repeated by 7.0⁻¹⁰ for submarine topography out to a radial distance of 101.5 statute miles (166.7 Violometers). Correction cartering ear water with denarty of 1.90 9/20⁻¹⁰ (see Bayer and Corbato, 1972). Vertical gravity gradient correction expressed in 9/20⁻¹⁰ for submarine topography out to a radial distance of 103.5 statute miles (166.7 Vertical gravity gradient correction expressed in 9/20⁻¹⁰ for lateral density of 1.90 9/20⁻¹⁰ (see Bayer and Corbato, 1972). Vertical gravity gradient correction expressed in 9/20⁻¹⁰ for lateral density changes (assumed to be two-dimensional) caused by geologic facies changes across the reset (see Figure 2.4). BHG density in 9/20⁻¹⁰ (see Figure 2.7). BHG density in 9/20⁻¹⁰ after correction for submarine topography and lateral density variations due to geologic facies changes and cratering Column 6

Column 7

processis. The of the set of the Column 8

Column 9 Column 10 Column 11

Average gamma-gamma density for depth interval in y/cm³. Gamma-gamma porosity in percent calculated from gamma-gamma density (column 12), grain density (column 8) and sea-vater density of 1.03 g/cm³. Average neutron porosity for depth interval in percent. Column 12 Column 13 Column 14

TABLE 2-5 (On opposite page). -- Bulk density, porosity, and grain density obtained from borehole gravity, gamma-gamma, and neutron measurements in borehole ORT-20 and from analysis of cores taken from ORT-20. Gamma-gamma and neutron data averages over depth intervals of BHG survey. Grain densities were calculated by the procedure described in Appendix 2-1 using data from Tremba and Ristvet (1986) and Ristvet and Tremba (1986).

-	2	٣	4	ъ	e '	7	80	6	10	11	12	13	14
136160.	1.693	.013	.130	.020	0.005	1.848	2.84	. 02	54.8	0.7	1.98	47.5	0.
160186.	1.857	.012	.133	.020	0.	2.010	2.86	.02	46.4	0.7	1.99	47.5	.0
186211.	1.673	.007	.134	.021	002	1.826	2.84	.02	56.0	0.4	1.96	48.6	
211226.	1.818	•	.135	.021	003	1.971	2.86	.02	48.6	.0	1.99	47.5	.0
226256.	1.772	•	.136	.021	003	1.926	2.85	.02	50.8	.0	2.00	46.7	.0
256271.	1.653		.137	.021	004	1.807	2.82	.02	56.6	.0	1.97	47.5	.0
271301.	1.787	.004	.138	.021	004	1.942	2.82	.02	49.1	0.2	.0	0.	
301331.	1.766	.002	.139	.021	004	1.922	2.78	.02	49.0	0.1		.0	.0
331361.	1.792	.007	.139	.022	005	1.948	2.88	.02	50.4	0.4	.0	.0	.0
361391.	1.833	.010	.139	.022	005	1.989	2.83	.02	46.7	0.6	2.04	43.9	
391421.	1.846	.014	.140	.022	005	2.003	2.75	.02	43.4	0.8	2.04	41.3	0.
421451.	1.857	.012	.140	.022	005	2.014	2.70	02	41.1	0.7	1.95	44.9	.0
451471.	1.891	.008	.140	.022	005	2.048	2.70	.02	39.0	0.5	1.90	47.9	0
471491.	1.852	.017	.140	.022	005	2.009	2.70	.02	41.4	1.0	2.06	38.3	0
491521.	1.792	.009	.140	.022	005	1.949	2.69	.02	44.6	0.5	1.93	45.8	0
521551.	1.770	.010	.140	.023	004	1.929	2.70	.02	46.2	0.6	1.91	47.3	<u>.</u>

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Depth interval in feet below sea level Apparent BHG density in g/cm³ (corrected for instrument calibration_,and drift and earth tides). Column

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standard deviation of repeated Ag/12 measurement expressed in g/cm³. Standard deviation of repeated Ag/12 measurements expressed in g/cm³. Vertical gravity gradient correction (expressed in g/cm³) for submarine topography out to a radial distance of 103.5 statute miles (166.7 Vertical gravity gradient correction expressed in g/cm³ for lateral density changes (assumed to be two-dimensional) caused by geologic facies vertical gravity gradient correction expressed in g/cm³ for lateral density changes (assumed to be two-dimensional) caused by geologic facies vertical gravity gradient correction expressed in g/cm³ for lateral density changes caused by crater-related processes and assumed to be vertical gravity gradient correction (expressed in g/cm³) for lateral density changes caused by crater-related processes and assumed to be symmetrical about Og2-18 (see Figure 2.7). BuG density in g/cm³ after correction for submarine topography and lateral density valiations due to geologic facies changes and cratering ŝ Column

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processes. ~ Column

 μ and grain density for depth interval in g/cm^3 (calculated from anneral and organic content percentages estimated from x-ray diffraction and lower-on-ignition analyses (see Appendix 2.2). Estimated uncertainty in mean grain density in g/cm³. BMG porosity in percent calculated from BMG density (column 8) and see-water density of 1.03 g/cm³. Uncertainty in 3MG porosity in porosity percent (calculated from standard deviation of BMG density (column 4) and uncertainty in mean grain Column 8

Column

• 5 Column

density (column 8). Awerage gamma-gamma-gamma-gamma-gamma-gamma-gamma-gamma density (column 12), grain density (column 8) and sea-water density of 1.03 g/cm³. Average neutron porosity in the second in percent. Column 11

Column 12 Culumn 13 Column 14 Culumn Column

TABLE 2-5.

TABLE 2-6 (On opposite page). -- Bulk density, porosity, and grain density obtained from borehole gravity, gamma-gamma, and neutron measurements in borehole OSR-21 and from analysis of cores taken from OSR-21. Gamma-gamma and neutron data averages over depth intervals of BHG survey. Grain densities were calculated by the procedure described in Appendix 2-1 using data from Tremba and Ristvet (1986) and Ristvet and Tremba (1986).

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TABLE 2-6.

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TABLE 2-7 (On opposite page). -- Bulk density, porosity, and grain density obtained from borehole gravity, gamma-gamma, and neutron measurements in borehole OTG-23 and from analysis of cores taken from OTG-23. Gamma-gamma and neutron data averages over depth intervals of BHG survey. Grain densities were calculated by the procedure described in Appendix 2-1 using data from Tremba and Ristvet (1986) and Ristvet and Tremba (1986).

14		.0	.0		.0	0			.0	.0	.7 facies be be n and rain g/cm ³ .	
13		0.	0.	.0	0.	.0	.0	.0	.0	0.	e miles (166 by geologic d assumed to and crater y diffractio ; ity in mean g ity of 1.03	
12		.0	.0		.0	.0		0	.0	.0	1.5 statut 2). statut 2). caused ocesses an ocesses an res change from x-ra from x-ra uncertain uncertain	
Ξ	0.2	0.5	0.7	0.5	0.7	0.4	0.3	0.5	1.0	0.8	ance of 10 thato, 197 thato, 197 thetafona related pr estimated estimated anaity of and sea	
10	45.7 44.7	46.1	46.7	45.8	43.7	42.8	35.7	41.4	41.5	39.1	1. 	
6	.04 .04	.04	.04	.02	.02	.02	.02	.02	.02	.02	irth tides upt to a rule (assumed i (assumed i assumed i	
8	2.85 2.85	2.85	2.85	2.85	2.84	2.74	2.72	2.73	2.81	2.88	iift and ca opoyraphy c f 1.90 y/cm f 1.90 y/cm sity changes sity changes aty (colum deviation deviation mn 12), gra	
7	2.018 2.037	2.011	2.000	2.017	2.049	2.008	2.117	2.026	2. 12	2.156	bration and d in g/cmal. c submarine t c submarine t t theral dens hy and lateral den hy and lateral a rom mineral a from standard from standard	
9	0.019 0.018	0.018	0.018	0.018	0.018	0.015	0.019	0.017	0.014	0.019	trument call a expressed in g/cm ³) for in g/cm ³) for in g/cm ³) for in g/cm ³) for the topograph calculated f calculated f calculated f calculated f rest.	
ניז	.022	.022	.022	.022	.022	.022	.023	.023	.023	.024	ed for ins essurement: expressed expressed in 9/cm ³ (c in 9/cm ³ (c in 9/cm ³ (c in 9/cm ³ (c in 10 9/cm ³ (c in 10 9/cm ³ (c in 21).	
4	.108	.111	.112	.113	.114	.116	.117	.119	.120	.122	<pre>>>>>>>>>>>>>>>>>>>>>>>>>>>>>>>>>>>></pre>	
n	.003	.010	.013	.010	.012	.006	.005	.008	.018	.015	In feet below and of repeations of gradient of gradient of gradient of gradient of to $Q2^{-18}$ (sr analyses (analyses (analyses (analyses (analyses (analyses (analyses (g)).	
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TABLE 2-7.

2..51

Table 2-8. -- Density model for atoll material surrounding OAK crater. Density layers are averages of BHG densities from OOR-17 and OSR-21 and are contrasted with the crater density model of Figure 2-7. Averaged grain densities and BHG porosities are also shown.

Approximate Depth Interval (fect below sea level)	Averaged BHG Density (g/cm ³)	Averaged Grain Density (g/cm ³)	Averaged BHG Porosity (%)
134 - 410	1.92	2.81	50
410 - 587	1.98	2.73	44
587 - 747	2.01	2.84	46
750 - 962	2.09 + .03 (es	timate from gamma-g	amma log run in
		OOR-17)	

ţ Table 2-9. -- BHG density and porosity values and their contrasts with respect reference borehole values (Table 2-8) for the averaged large intervals shown in Figures 2-16 and 2-19.

0PZ-18			2.13(+.16) 36(-9)		2.11(+.13) 37(-7)		2.05(+.04) 44(-2)		92-94
0TG-23			2.05(+.08) 2 41(~4)	ge H/I	2.03(+.05) 2 40(-4)		11		
<u>00</u> T-19	2 4 1.93(+.01) 50(0)	90	2.00(+.03) 44(-1)	o facies chan	2.00(+.02) 43(-1)	to J/K	2.03(+.02) 45(-1)		95-97
<u>OTG-20</u>	ntinuity 1 to 1.92(0) 51(+1)	ntinuity 4 to	1.98(+.01) 44(-1)	nformity 5 to	1.97(01) 44(0)	s change H/I	11		
17/0SR-21	rom discor 1.92 50	rom disco	1.97 45	from disco	1.98 44	from facie	2.01 46		100
00R-1	<u>Interval approximately f</u> Density(contrast) in g/cm ³ Porosity(contrast) in %	Interval approximately f	Density(contrast) in g/cm ³ Porosity(contrast) in %	Interval approximately f	Density(contrast) in g/cm ³ Porcsity(contrast) in %	Interval approximately f	Density(contrast) in g/cm ³ Porosity(contrast) in %	Mass columns (%)	1

APPENDIX 2-1

BOREHOLE GRAVITY SURVEY, BOREHOLE E-1, MEDREN ISLAND

The BHG survey in borehole E-1 on Medren (ELMER) Island (see fig. 2-20) was conducted in April, 1984, by the U.S. Geological Survey to determine if reliable BHG data could be gathered in the microseismic environment of an atoll and to evaluate the range of natural density variations of reef-forming materials. Near-surface vibrations caused by wave action were minimal and the repeatability of BHG measurements generally was excellent. The tabulated data for this survey are given in Beyer, Ristvet, and Oberste-Lehn (1986).

The borehole gravity survey in the E-1 borehole shows that the bulk density of atoll materials to a depth of 1,800 ft ranges from about 1.9 to about 2.3 g/cm³ and averages slightly more than 2.0 g/cm³ at the scale examined by the BHG survey. Several density patterns are evident.

- 1. Higher densities between 1,140 and about 1,290 ft correspond to harder rocks as indicated by slower drill rates (fig. 2-20).
- 2. The gravity station at 1,410 ft (point labeled "A" in fig. 2-20) probably is in close proximity to a sizable cavern that has caused measured gravity to be unexpectedly low. This one anomalous gravity reading incorporated into the overlying and underlying density calculations explains the generally high and low densities of the two adjacent intervals.
- 3. A repeated pattern of density variations (labeled "l" through "5" in fig. 2-20) may be due to facies changes and/or diagenesis associated with relative sea-level changes. These repeated patterns of downward decrease in density (increase in porosity) followed by more abrupt increase in density (decrease in porosity) should be examined for possible correlation with available geologic data.

Densities in the upper 600 ft are slightly higher than the densities over the same depth interval in PEACE Program reference boreholes OOR-17 and OSR-21 at OAK crater. Part of this may be due to the E-1 borehole being much closer to the ocean edge of the reef than are OOR-17 and OSR-21. Boreholes OOR-17 and OSR-21 are more likely to be in a less dense, more lagoonward facies. Corrections for submarine topography are more critical at the E-1 borehole because of its closer proximity to the outer reef slope than the PEACE Program boreholes. Unfortunately, bathymetry is less well known adjacent to the E-1 borehole, and some of the density differences between E-1 and OOR-17 and OSR-21 may be due to errors in corrections for submarine topography at E-1.



FIGURE 2-20 - BHG density profile for borehole E-1, Medren (ELMER) Island. Drilling time profile and geologic ages are from Ladd and Schlanger (1960). Large interval averages of density along righthand depth scale correspond to vertical dashed lines. Diagonal dotted lines labeled "1" through "5" designate suggested repeated density (porosity?) cycles.

APPENDIX 2-2

DETERMINATION OF INTERVAL GRAIN DENSITY

An estimate of interval grain density is needed before BHG porosity can be calculated from BHG density. Grain densities of individual core samples were estimated from x-ray mineralogy and organic analyses by Tremba and Ristvet (1986) and Ristvet and Tremba (1986) and are shown in Figures 2-8 through 2-13.

An example of how grain density was calculated from x-ray mineralogy and organic analyses follows. Calcite (and magnesium calcite), aragonite, and organic matter were assigned grain densities of 2.72, 2.93, and 1.00 g/cm³ respectively. If organic matter was present and measured in weight percent of dry solids (generally 3 percent or less), the remainder of the dry sample was assumed to consist of inorganic material (generally 97 percent or more). Thus, for a sample with the analysis

Calcite	Aragonite	Organic Matter
(wt %)	(wt %)	(wt %)
29	71	2.5

The grain density is

 $[(.29)(2.72) + (.71)(2.93)](1-.025) + .025 = 2.82 \text{ g/cm}^3$

If the sample had no measurable organic matter, the grain density is

 $(.29)(2.72) + (.71)(2.93) = 2.87 \text{ g/cm}^3$

The plots of grain densities of core samples were generalized to average grain densities for BHG intervals as shown in Figures 2-8 through 2-13. Grain densities averaged by sedimentary packages by Tremba and Ristvet (1986) were too generalized for the BHG data. Uncertainties of \pm .02 or \pm .04 g/cm³ were assigned in order to estimate errors in porosity calculations (columns 8, 9, 11 of Tables 2-2 through 2-7; also see Appendix 8-2, Beyer, Ristvet, and Oberste-Lehn, 1986).

CHAPTER 3:

PALEONTOLOGIC EVIDENCE FOR SEDIMENTARY MIXING IN OAK CRATER

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Thomas M. Cronin and Thomas G. Gibson¹

INTRODUCTION

In 1985, during the course of paleontologic studies of OAK and KOA craters, Enewetak Atoll, it was discovered that the analysis of the distribution of microfossils aided the understanding of the dynamic processes involved in the formation and evolution of the nuclear craters (Cronin, Brouwers, and others, 1986; Brouwers, Cronin, and Gibson, 1986; Henry, Wardlaw, and others, 1986; Henry and Wardlaw, 1986; and Wardlaw and Henry, 1986). These paleontologic studies were particularly useful in determining the depth of origin (or provenance) and sedimentologic history of the disturbed and mobilized materials that partially infilled KOA crater after the initial excavation by the detonation of the nuclear device.

The primary purpose of the present study is to determine the composition and provenance of crater-fill materials and the nature of sediment mixing in OAK crater using micropaleontologic data. This study is an extension of the paleontologic component by the U.S. Geological Survey for the PEACE Program (Cronin, Brouwers, and others, 1986; Brouwers, Cronin, and Gibson, 1986). In this study of OAK, we intend to establish the depth limits of mixing of: (1) surficial material, (2) sediment from the uppermost 50 ft of the stratigraphic section, and (3) material from intermediate depths (50 to 300 ft). Furthermore, we intend to determine the pattern of "piping" of deep material emplaced in the crater-fill from horizons 500 to 900 ft below the lagoon bottom or sea floor. Our results are integrated with geologic and geophysical data to form a general model of crater formation in Chapter 7 of this Open-File Report.

MATERIAL AND METHODS

To accomplish our objectives, detailed restudy of samples from reference boreholes OAR-2A and OOR-17 was necessary to refine our zonation of the microfaunal sequence in the upper 400 ft of the stratigraphic section (for discussion of the succession of microfaunal zones used on Enewetak, see Cronin, Brouwers, and others, 1986, and Brouwers, Cronin, and Gibson, 1986). The laboratory and biostratigraphic procedures used herein are the same as those described in the reports cited above. Microfossils were extracted from sediment between 63 and 850 µm in grain size. Samples from two central-crater (ground-zero) boreholes (OBZ-4 and OPZ-18) and three transition boreholes

¹ Branch of Paleontology and Stratigraphy, U.S. Geological Survey, Reston, VA 22092. (OCT-5, OFT-8, and OKT-13) were examined in detail for the mixing study of OAK crater. Table 3-1 lists the depths of all 159 samples studied.

Throughout this report, depths of zonal boundaries are occasionally rounded off to whole numbers for convenience. Of course, the accuracy of any particular faunal zone is limited by the resolution of the sampling interval.

STANDARD MICROFAUNAL SEQUENCE

Quantitative data on the occurrence of diagnostic ostracode species (Appendices 3-1 and 3-2) and semiquantitative data on benthic foraminifers (Gibson and Hill, <u>in preparation</u>) from boreholes OAR-2/2A and OOR-17 were used to improve the standard zonation of Cronin, Brouwers, and others (1986) and Brouwers, Cronin, and Gibson (1986), in which 12 faunal zones, designated AA through MM, in descending order, were defined. In addition, the percent of specimens of the ostracode <u>Neonesidea schulzi</u> with preserved setae¹ was used as a new measure to quantify the amount of material mixed downward from the surface. Only living or recently dead specimens of this species found in surficial lagoon sediments have setae preserved (generally 70 to 80%). Ostracode setae normally are degraded and destroyed by natural processes soon after death of the organism and burial of the shell. Therefore, the occurrence of setae in specimens below the sediment surface in the crater-fill materials is taken to indicate mixing of specimens from the surface sediments.

The following zones were used in the quantitative analyses of ostracodes:

- Surface: The percent of Neonesidea schulzi with setae preserved.
- Zone AA: The combined percentages of <u>Hermanites mooneyi</u> and Loxoconchella sp. A.
- Zone BB-CC: The combined percentages of <u>Cletocythereis</u> sp. A and <u>Loxoconcha heronislandensis</u>.
- Zone EE-FF: The combined percentages of <u>Caudites</u> sp. A, <u>Caudites</u> sp. B, <u>Cletocythereis</u> rastromarginata, <u>Loxonconcha</u> labrynthica, and Loxoconchella sp. C.
- Zone FF-GG: The combined percentages of <u>Australimoosella</u> sp. A, <u>Bythoceratina</u> sp. A, <u>Cletocythereis canaliculata</u>, <u>Procythereis sp. A</u>, and <u>Semicytherura</u> sp. A.
- Zone II-MM: The combined percentages of all species restricted to zones II, JJ, KK, LL, and MM as determined by Cronin, Brouwers, and Gibson (1986). In Appendices 3-1 and 3-2 at the end of this Chapter, the totals for these species are given in row 41. Procythereis sp. B generally occurs in zones

 $\frac{1}{2}$ Setae are small hairs that occur on the exterior of the values of some taxa of ostracodes.

OAR-2A	0BZ-4	0CT-5	OFT-8	OKT-13	00R-17	OPZ-18
0.25	2.8	0.2	8.75	10.4	0.25	7.0
2.3	11.8	8.8	18.6	18.5	14.15	35. 0
6.0	21.1	17.5	27.9	25.4	25.75	44.6
9.3	33.0	39.5	35.1	28.75	38.4	57.85
11.85	40.5	57.55	43.1	36.0	49.7	74.3
14.5	58.5	66.8	48.85	55.65	60.2	89.45
17.1	66.35	76.65	64.0	59.9	66.75	102.0
20.8	75.15	86.15	74.0	68.2	72.8	115.05
22.75	84.15	95.35	-	80.0	83.7	131.0
23.75	93.1	104.25	-	-	89.8	139.7
26.05	104.55	113.15	-	-	100.45	154.2
31.9	112.9	124.0	-	-	101.4	169.35
34.75	121.8	132.8	-	-	110.5	174.95
40.2	130.0	140 .9	-	-	119.1	182.3
43.55	144.5	149.65	-	-	125.25	198.0
62.5	151.55	157.6	-	-	131.95	207.3
74.8	166.85	166.4	-	-	137.15	210.4
90.4	178.6	176.25	-	_	146.1	229.95
95.8	186.8	186.0	-	_	154.25	232.1
115.1	193.6	_	-	_	165.6	239,15
127.8	196.5	-	-	-	173.05	_
134.0	205.1	-	_	-	184.25	-
157.45	213.9	-	-	-	193.6	-
171.2	225.65	-	-	_	200.8	_
188.25		-	-	-	209.3	-
195.3	-	-	_	_	215.5	-
204.9	-	_	-	-	226.05	-
212.45	-	-	_	_	233.35	_
223.9	_	-	-	_	239.0	_
234.6	-	-	-	_	250.3	_
244 55	_	-	-	_	250.5	_
244.55	_	_	-	_	270 1	_
240.0	_	_	-	_	270.1	_
200.40	_	_	_	_	203.05	_
202.05	_	_	_		272.1	_
203.7	_	-	_	-	299.15	-
370 5	-	_	-	-	310.1	-
5/905	-	-	-	-	320.2	-
-	_	-	-	-	331.2	-
-	-	-	-	-	339.0	-
-	-	-	-		36/.9	-

TABLE 3-1. — Depth (ft bsf) in boreholes of samples examined during the study of the mixing of crater-fill materials from OAK.

Ν

II-MM; however, it does occur higher in the section in single samples from OOR-17 (331.2 ft bsf) and in OAR-2A (223.9 ft). Specimens of <u>Procythereis</u> sp. B in crater boreholes are considered piped, so that the percent of II-MM species includes species groups 36 and 41 from the appendices.

For the companion analysis of the foraminifers, the zones used are characterized as follows:

- Surface: The presence of chitinous inner linings and original (natural) coloration in several species.
- Zone AA: The presence of <u>Calcarina spengleri</u> and <u>C. hispida</u>. Upper AA is characterized by coloration in specimens of <u>C</u>. <u>spengleri</u> that is not present in specimens in the lower part of this zone, as determined from the reference boreholes.

Zone BB: The presence of Epistominella tubulosa and Anomolina sp. A.

Zone CC: The presence of advanced forms of Calcarina rustica.

Zone EE: The presence of <u>Calcarina delicata</u> and primitive forms of Calcarina rustica.

Zone FF-GG: The presence of Calcarina calcar and Cibicides sp. A.

The percentages of ostracodes for each category (excluding the Surface¹ and II-MM categories) are plotted for boreholes OAR-2A and OOR-17 in Figures 3-1 and 3-2. These were used for comparison with the mixed faunal sequences in the central crater and transition boreholes. It is noteworthy that the two faunal sequences in OOR-17 and OAR-2A are very similar to each other, enhancing the accuracy of estimates of the original depths of mixed specimens.

A large proportion (generally about 40 to 60% of each sample) consists of long-ranging species not restricted to a particular zone. Some of these nondiagnostic species probably also were mixed during crater filling. Consequently, the percentage values for samples from crater boreholes are, in some cases, <u>minimum</u> values (i.e., if mixed specimens of non-diagnostic species could be identified, the true percentage of an assemblage from any particular zone would be slightly higher).

In many ways, the use in the mixing study of selected species that have acme zones (intervals of greatest abundance) is an exercise in probability. Those species chosen as diagnostic of zones AA, BB-CC, EE-FF, and FF-GG in the upper 400 ft of normal stratigraphic section have a high probability (generally about 80 to 90%) that they originated from within that interval.

¹ Sediments from the lagoon floor or the upper several inches of sediment below the lagoon floor itself are referred hereafter as Surface materials.







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Confirmation of a zone assignment from both fossil groups increases significantly the probability that the zone assignment is correct.

The use of percentages of specimens from zones II-MM probably underestimates the real percentages by no more than 5 to 10%, because far fewer species from the upper 300 ft of section range downward into these zones. This index of piped specimens is considered an accurate indicator of the proportion of piped material represented in a sample. The percentage piped from a depth interval may be considered representative of the entire sedimentary assemblage at that level if two assumptions are correct. First, we must assume all particles of all sizes behave the same as those between 63 and 850 µm (the size range from which ostracodes were extracted). Second, we must assume sediment particles of different shapes and densities behave the same as the ostracode valves and carapaces. With these assumptions in mind, and given the error margins associated with the limits to microfossils zonations discussed above, these data are useful in making volumetric estimates of the proportion of crater-fill sediments piped from depth.

The percentage of <u>Neonesidea schulzi</u> having setae is a distinct type of index that gives an approximate estimate of the actual percentage of surface material, at least to the extent that it can be determined from using this one common species of ostracodes.

A large proportion of the foraminifer assemblage in most samples is composed of <u>Amphistigina madagascarensis</u>. This species is abundant in the modern reef environments on Enewetak and continues downward into the Miocene strata in the Enewetak boreholes. Therefore, its occurrence alone cannot be used for biostratigraphic determination; however, its preservation state is indicative of its zone of origin or provenance. Translucent specimens of this species occur only in zone AA. Below this zone (i.e., in the Pleistocene section and in older strata), they are opaque. Thus, the occurrence of translucent specimens of <u>A. madagascarensis</u> indicates that their provenance is the Holocene section (zone AA). Many other foraminifer species also have long ranges and cannot be placed definitely. However, the evolutionary changes in the <u>Calcarina</u> lineage are most helpful for determination of the horizons, particularly because they are among the most numerous species in the assemblage.

In some other cases, the preservation of ostracodes and foraminifers also is important in identifying provenance. For example, conspicuous brown specimens of long-ranging species clearly could be identified as originating from deeper zones II-MM. Also, in the injection dikes between 189 and 208 ft bsf and at 233 ft bsf in OPZ-18, the preservation state is almost identical to that of specimens in the upper part of zone AA; therefore, the origin of even non-diagnostic species with the appropriate shell preservation can be shown confidently to be from zone AA.

RESULTS

The following results can be shown from our current studies of samples from OAK crater.

Central Crater (Ground Zero) Boreholes

Boreholes OBZ-4 and OPZ-18 were cored near ground zero in OAK crater; we examined 39 and 21 samples, respectively, from each. The following is an informal zonation of the crater-fill materials based on the characteristics of the mixing of microfossils. The zones of material in the crater-fill from top to bottom are: (1) the **Homogenized Zone**, (2) the **Upwardly Mixed Zone**, (3) the **Maximum Piping Zone**, and (4) the **Basal Mixed Zone**. The boundaries between these zones are gradational and their depths approximate. In addition, we examined material from several injection dikes. The results are based on the ostracode-occurrence data given in Appendices 3-3 and 3-4, many of which are presented graphically in Figures 3-3 and 3-4, and the benthic foraminifer data is summarized in Tables 3-2 through 3-6, located at the end of this Chapter immediately preceeding the Appendices. To appreciate the nature of the mixing described in the next few pages and to see the actual percentage values, it is useful to compare directly the "normal" pattern of ostracodes (figs. 3-1 and 3-2) with that of the mixed sequence (figs. 3-3 and 3-4).

Homogenized Zone (0 to 40 ft). -- In this interval, high percentages (50 to 60%) of Neonesidea schulzi with preserved setae and specimens of Discorbis and Cymbaloporetta with chitinous inner linings originated from the Surface. High percentages of AA species, low to moderate numbers of specimens from CC, low to moderate occurrences of EE-GG species, low percentages of presumably piped specimens of II-MM species, and 3 to 6% BB-CC mixed material also characterize the Homogenized Zone. In general, this interval is easily identified by its anomolously high species diversity, resulting from the homogenization of material from virtually all zones with apparently equal contributions from most sub-AA zones. Specimens from the Homogenized Zone are characterized by widely varying preservation states.

Upwardly Mixed Zone (40 to 100 ft). -- This interval coutains consistently low percentages of EE-GG ostracodes and greater percentages of piped material from zones II-MM than occur in the upper 40 ft of OAK craterfill. The absence of surface material is conspicuous (with the exception of a single sample from 84 ft bsf from OBZ-4). Some samples from the Upwardly Mixed Zone contain less AA material than the overlying Homogenized Zone; in others, zone AA foraminifers still predominate. This interval characteristically contains moderate amounts of BB-CC material. The boundary between the Homogenized Zone and the underlying Maximum Piping Zone is not sharp, although this may be due to sample spacing. However, the relative contributions to the Upwardly Mixed Zone of foraminifers and ostracodes from various zones are quite distinct from the rest of the crater- fill. The piped specimens from this zone are from KK-LL and possibly from MM.

Maximum Piping Zone (100 to 160 ft). -- The highest percentages of piped specimens (9 to 12% in both OBZ-4 and OPZ-18) occur in this zone. LL-MM zone foraminifers and ostracodes are common at 121.8 ft bsf in OBZ-4, where at least eight separate ostracode species were emplaced from depth. Low percentages of AA foraminifers are characteristic of the upper part of the Maximum Piping Zone; however, no definite AA ostracodes or foraminifers are recorded from below about 125 ft in OBZ-4. This interval contains low to moderate numbers of specimens from zones CC and EE-GG. Here, piped foraminifers are not as obvious in OPZ-18 as in OBZ-4. Anomolously large numbers of single ostracode valves and still-articulated carapaces are broken



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in samples from the Maximum Piping Zone, suggesting a kind of shock fracturing. The base of the Maximum Piping Zone is marked by an abrupt drop in the percentage of piped specimens in the samples.

Basal Mixed Zone (160 to 190 ft). -- The Basal Mixed Zone contains low percentages of piped material, low percentages of zone AA foraminifers, and high percentages of zone EE-FF material. Most of the sediment in the Basal Mixed Zone probably originated from zones EE and FF.

Statistical Analysis of Crater-Fill Materials. -- A simple linearregression analysis of depth versus percentage of piped specimens was performed for samples from the upper 160 ft of OBZ-4 and the upper 145 ft of OPZ-18 (i.e., for all of the samples taken from above the Basal Mixed Zone in both boreholes). This statistical analysis was conducted to further analyze piping in the crater-fill from boreholes OBZ-4 and OPZ-18. Figures 3-5a and 3-5b show this relationship for 20 upper samples from OBZ-4 and 14 samples from OPZ-19. A positive correlation exists with correlation coefficients of r = 0.46 and r = 0.52, respectively. If samples from these depth intervals containing no piped specimens are excluded (9 samples in OBZ-4, 1 in OPZ-18; see Appendices 3-3 and 3-4), the correlation coeffecients are much higher, r =0.64 and r = 0.93, respectively (figs. 3-5c and 3-5d). The absence of piped specimens in some samples may be a result of the small number of specimens that could be extracted. Nonetheless, in both boreholes there is a positive correlation, suggesting a diminishing contribution of piped material toward the upper intervals of crater-fill.

Injection Dikes. -- Injection dikes were sampled only in borehole OPZ-18 from 189 to 208 ft and 233 ft bsf. Well-preserved AA foraminifers and ostracodes and many articulated, translucent ostracode carapaces occur in these samples. Bright-red Homotrema is also common. The samples from 189.25, 198.0, and 207.3 ft bsf are composed of almost identical assemblages of species, and the preservation is almost identical also. In these dikes, material from BB-CC is conspicuously missing. All evidence suggests an origin for almost all material between 189 and 208 ft from the upper part of zone AA; however, the lack of <u>Neonesidea schulzi</u> with setae argues against any material originating from the Surface. The sample at 210.4 ft contains recrystallized microfossils, and the samples at 229.95 and 232.1 ft bsf contain zone AA species. These are mixed with zone EE-GG species. No piped II-MM zone material occurs in this dike.

Transition Boreholes

Transition boreholes OCT-5, OFT-8, and OKT-13 were sampled for microfossils for the current mixing study.

OCT-5. -- Samples from borehole OCT-5 were analyzed semiquantitatively for ostracodes (Appendix 3-5) and foraminifers. Both microfossil groups, particularly ostracodes, are much less abundant than in samples from OBZ-4 and OPZ-18, and the following zonation is based more heavily on the foraminifers.

0-9 ft. -- Samples from this interval in OCT-5 have anomolously high diversity, with approximately equal contributions from all mid-upper zones (AA-GG), as was the case in the Homogenized Zone of OBZ-4 and OPZ-18. A few piped toraminiters are found at 8.8 ft bsf.



having no specimens borehole OB2-4, upper 160 160 ft, samples upper 145 ft, samples having no piped borehole OBZ-4, upper (23) Figure 3-5. -- Plot of percentages of piped specimens versus depth. borehole OPZ-18, borehole OPZ-18, upper 145 ft; (5c) (PS) piped specimens omitted; omitted. ft; (5b)

PIPED SPECIMENS VS DEPTH

39-77 ft. -- Abundant zone AA foraminifers, sparse BB-CC ostracodes, and low percentages of EE-GG material are found in samples from these depths.

86-105 ft. -- Samples from here differ from the overlying ones in lacking zone AA species; this interval has mostly CC-GG foraminifers; however both ostracodes and foraminifers are extremely sparse.

113-140 ft. -- Zone AA species predominate in these samples; preservation of the specimens is similar to specimens from zone AA and bright-red Homotrema (indicative of zone AA) also occurs; samples from here contain low percentages of CC-GG foraminifers.

149-187 ft. -- In this borehole, samples from this interval are almost barren and contain no diagnostic species of either ostracodes or foraminifers. Part of the explanation for the paucity of ostracodes may be that zone EE (normally at roughly comparable depths in the reference boreholes) typically contains few ostracodes in the normal stratigraphic section. However, samples from 149 to 187 ft in OCT-5 also lack even benthic foraminifers, which do occur in zone EE.

OFT-8. -- This borehole has diagnostic microfaunas in all samples examined, allowing a threefold subdivision of the upper 75 ft (Appendix 3-6).

0-19 ft.-- This is a mixed interval containing material from zones AA to probably no deeper than FF. Samples from this interval resemble the Homogenized Zone of the upper parts of other central crater and transitional boreholes. In the uppermost sample at 0.0 to 0.25 ft bsf, a single foraminifer and a single ostracode specimen occur, suggesting piping from zones JJ-MM.

27-50 ft. -- This interval consists almost entirely of material from zone AA, as indicated by the foraminifers. The ostracode species also occur typically in AA and are preserved like those from that zone. There is a noticable absence of zone BB-CC material, also indicating a lack of mixing.

64-75 ft. -- At 64 ft bsf, a mixture of AA and sparse FF-GG foraminifers occurs with typical BB-CC ostracodes. The 74.0-ft sample appears to be from sediment that is essentially in place and consists exclusively of BB-CC material. A detailed sampling across the interval from 50 to 75 ft would be necessary to better document the transition into undisturbed sediments at this borehole site.

OKT-18. -- This borehole contained highly diagnostic ostracodes and foraminifers that allowed a fourfold subdivision of the upper 80 ft (Appendix 3-11). The results from the two fossil groups match each other more consistently, aud, thus, these zones are more definitive than in any borehole yet analyzed.

0-19 ft. -- Samples from this interval are noted for their anomolously high species diversity and homogenization of zone AA-GG material. These samples resemble those from the uppermost parts of borcholes 052 4, 072 18, and 0CT-5. No piped specimens are found.

25-37 ft. -- These samples contained almost exclusively material from zone AA; small percentages from EE-GG are noted from the sample at 25.4 ft

bsf. The microfaunas from this interval resemble those from the injection dike in OPZ-18 in both species composition and preservation.

55-66 ft. -- The samples studied contain only material from zones CC-DD. Especially noteworthy is the occurrence of Paracytheridea remanei (which has its acme in DD in all reference boreholes) in OKT-13 at 55.65 (abundant), 59.9, and 68.3 ft. Also, the abundance of Orionina sp. at 59.9 ft is noteworthy. This latter very distinctive species is abundant in OAR-2A at 62 to 75 ft, and a biostratigraphic correlation is probable for strata between 59.9 ft in OKT-13 and 62-75 ft in borehole OAR-2A.

68-81 ft. -- A typical EE-FF assemblage occurs in this interval; there is no obvious mixing from AA or BB.

SUMMARY AND CONCLUSIONS

Our primary conclusions from the mixing study for the OAK crater area follow:

- 1. <u>Piped material</u>: an inverse relationship exists between sample depth and the percentage of piped material (from zones II-MM) in OBZ-4 between the surface and 160 ft bsf and in OPZ-18 between the surface and 145 ft bsf. Sparse piped specimens occur in the upper 10 ft of OCT-5 and the upper 1 ft or OFT-8; no piped specimens were found in OKT-13.
- Mixing of abundant <u>Surface material</u> occurs in OBZ-4 and OPZ-18 downward to a depth of 35 ft, although sparse specimens from the Surface occur as deep as 84 ft in OBZ-4.
- Mixing of abundant material from zone AA is evident in OBZ-4 and OPZ-18 to about 50 to 60 ft bsf; AA material is less common to a depth of about 120 ft bsf in both boreholes.
- 4. Mixing of moderate amounts of <u>material from zones EE-GG</u> (occurring from 100 to 300 ft bsf in the normal stratigraphic sequence) is encountered in the upper 100 ft of the two central-crater boreholes (OBZ-4 and OPZ-18) and in the upper 20 ft of the transition boreholes. Mixing of BB-CC material is less significant than that of EE-GG material in all boreholes.
- 5. An apparent injection dike between 189 and 208 ft bsf in OFZ-18 contains almost exclusively AA microfossils. In addition, sediment from these dikes is greenish-gray, like that from the normal AA section. Distinctive microfaunas at 25 to 37 ft in OKT-13 and 27 to 50 ft in OFT-8 are extremely similar to those in this injected OPZ-18 material, although it is not clear if they are genetically related.
- 6. A <u>Homogenized Mixed Zone</u> containing approximately equal proportions of AA-GG material is a general characteristic of all central-clater (ground-zero) boreholes (down to 40 ft) and in transition boreholes (down to 20 ft).

7. The overall consistency between the ostracodes and foraminifers and our ability to quantitatively revise the standard and mixing zonations to a high degree of resolution gives us confidence that the only limits to our ability to further refine zonations of mixed material, to more accurately identify provenance, and to improve volumetric computations of mixed materials are manpower constraints and sample/core recovery.

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DEPTH (ft bsf)	FORAMINIFER DATA
2.8-3.05	Mostly AA, some mixing from CC-CG (most likely CC), piping from KK-MM.
11.8-12.05	Mostly AA, moderate amount of CC-GG (most likely CC), piping from JJ-KK.
21.1-21.35	Moderate amount of AA, some CC and EE-GG, piping from II-KK.
33.0-33.25	Mostly AA, mixed with minor CC.
40.05-40.3	Moderate amounts of AA, CC, and EE-GG, moderate amount of piping from II-LL.
58.50-58.75	Mixed AA, CC, and EE-GG, moderate amount of piping from KK- LL.
66.35-66.60	Mixed AA, CC, and EE-GG, more of ?CC or EE-GG than in above sample, some brown specimens presumably from KK-LL.
75.15-75.4	Mixed AA, CC, and EE-GG, more of EE-GG, some brown specimens presumably from KK-LL.
84.15-84.4	Mixed AA, CC, and EE-GG, some piping from KK-LL, possibly M1.
93.1-93.35	Mostly CC and EE-GG, sparse AA, some brown specimens presumably from KK-LL.
104.55-104.8	Mostly CC and EE-GG, sparse AA, piping from KK-LL.
112.9-113.15	No definite AA; some from ?CC, definite EE-GG, piping from KK-LL and possibly from MM.
121.8-122.05	Definite AA and CC and ?EE-GG, some piping from KK-LL.
130.0-130.25	Mostly CC with some EE-GG, sparse brown specimens possibly from deeper zones.
144.5-144.75	?CC and EE-GG.
151.55-151.8	?CC and EE-GG, some brown specimens presumably from deeper zones.
166.85-167.1	?CC and EE-GG.
178.6-178.85	<pre>?CC and EE-GG, some brown specimens presumably from deeper zones.</pre>
186.8-187.05	?CC and EE-GG.
193.6-193.85	?CC and EE-GG.
196.5-196.75	EE-GG (probably FF) with ?CC.
205.1-205.35	EE-GG (probably FF) with ?CC.
213.9-214.15	EE-GG.
225.65-225.9	EE-GG.

Table 3-2.--Summary of foraminifer occurrences in OAK crater borehole OB2-4.

DEPTH (ft bsf)	FORAMINIFER DATA
0.2-0.45	Specimens from EE-GG, abundant AA, CC; some specimens from II-MM.
8.80-9.05	Mostly specimens from EE-GG, some from AA, some from KK-MM.
17.5-17.75	Some EE-GG, probably BB-CC, no AA.
39.50-39.75	AA and EE-GG with possibly CC and possibly deep zones.
57.55-57.8	Abundant AA and EE-GG.
66.8-67.05	Abundant AA, probable CC and definite EE-GG.
76.65-76.9	Mostly AA, some EE-GG.
86.15-86.4	All from CC-GG, no AA.
95.35-95.6	Same as above sample.
104.25-104.5	Very few specimens, but similar to above sample.
113.15-113.4	Few specimens, definite AA dominant, some from CC.
124.0-124.25	Abundant AA, sparse specimens probably from CC, possibly CC- GG.
132.8-133.05	Almost entirely AA, few specimens from CC or CC-GG.
140.9-141.15	Same as above sample.
149.65-149.9	Barren.
159.6-157.85	Only I specimen probably from CC, possibly from BB-GG.
166.4-166.65	Only 1 specimen, provinence uncertain.
176.25-176.5	Only 1 specimen, provinence uncertain, probably CC-GG.
186.0-186.25	Few specimens of CC-GG, probably CC.

Table 3-3.--Summary of foraminifer occurrences in OAK crater borehole OCT-5.

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Table 3-4.--Summary of foraminifer occurrences in OAK crater borehole OFT-3.

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DEPTH (ft bsf)	FORAMINIFER DATA
0.0-0.25	Few specimens, definitely AA (but not uppermost AA), few specimens from EE-GG, scattered light-brown material with l immature specimen possibly from JJ-MM, not certain there
8.75-9.00	More abundant AA (with more from uppermost AA), sparse specimens from CC, moderate amount from EE-GG, no brown material.
18.60-18.85	Abundant specimens from upper AA, few specimens probably from EE-GG.
27.90-28.15	All specimens from upper part of AA.
35.1-35.35	All specimens from upper part of AA.
43.1-43.35	All specimens from AA, probably slightly lower AA than two samples above.
48.85-49.1	All specimens from AA.
64.0-64.25	Mixture of older part of AA with few from EE-GG.
74.0-74.25	Appears to be unmixed BB-CC.

DEPTH (ft bsf)	FORAMINIFER DATA
10.4-10.65	Mostly EE-GG, some from lower AA.
18.5-18.75	Mostly CC, some from EE-GG mixed with specimens from AA.
25.4-25.65	Specimens from ?CC mixed with few from EE-GG, more abundant AA than in above two samples
28.75-29.00	All from AA.
36.00-36.25	All from AA.
55.65-55.90	All apparently from CC-DD.
59.5-59.75	All probably from CC-DD.
68.2-68.45	All probably from EE-FF, possibly some CC-DD.
80.00-80.25	All from CC-GG.

Table 3-5.--Summary of foraminifer occurrences in OAK crater borehole OKT-13.

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DEPTH (ft bsf)	FORAMINIFER DATA
7.0-7.25	Mostly AA, with some CC and EE-GG mixed, has light-brown material but no diagnostic specimens.
35.0-35.25	Mostly AA, with some CC and EE-GG mixed, has more light-brown material than above sample but only one probable specimen from KK-MM.
44.6-44.85	AA, with some CC and more EE-GG than above, moderate amount of light-brown material but no diagnostic specimens.
57.85-58.1	More AA than above sample, with EE-GG common, moderate amount of light-brown material but no diagnostic specimens.
74.3-74.55	Mostly AA with some specimens from EE-GG, moderate amount of light-brown but no diagnostic specimens.
89.45-89.7	Mostly AA with some EE-GG, moderate amount of light-brown material with 1 specimen possibly from KK-LL-
102.0-102.25	Few specimens, some AA, some EE-GG, and moderate amount of light-brown material with one specimen questionably from LL-MM.
115.05-115.3	Few specimens, some AA, some EE-GG, moderate amount of light-
131.0-131.25	Few specimens, some AA, some EE-GG, 1 specimen possibly from KK-MM.
139.7-139.95	Some AA, some EE-GG, some light-brown material.
154.2-154.45	Some AA, some EE-GG, some light-brown material.
169.35-169.6	Some AA and some EE-GG, sparse light-brown material.
174.95-175.1	Some EE-GG, some probably AA, mostly uncertain.
182.3-182.55	Some AA, some EE-GG, very little light brown material.
189 25-189.5	All AA, apparently upper AA.
198.0-198.25	All AA, apparently upper AA.
207.3-207.55	All AA, apparently upper AA.
210.4-210.65	All recrystallized, but no markers??
229.95-230.2	No diagnostic species, most look like ?CC-GG, possibly few AA.
232.1-232.35	Mostly AA, probably upper A, except for 1 specimen from EE- GG.
239.15-239.4	Appearently all EE-GG, probably FF-GG.

Table 3-6.--Summary of foraminifer occurrences in OAK crater borehole OPZ-18.

BOREBOLE OAR-2/2A

									IJWVS	E DEP	TH (fi	: bsf)							
i	DIAGNOSTIC SPECIES	0.25	2.3	6.0	9.3	11.85	14.5	17.1	20.8	22.75	28.05	31.9	34.75	40.2	43.55	62.5	74.8	90.4	95.8
- -	Australimoosella sp.		1	,	,	.	,	,		.			'	1	.				1
،	Bythocerating sp.	ī	I	,	ı	,	ı	4	,	1	•		'	,	I	,	ı	1	,
÷.	Callistocythere sp. A	1	1	ł	1	ı	ı	,	9	5		'	4	1	I	4	-	,	1
4.	C. sp. B	I	1	٦	e	,	-1	2	I				+	• •	ı	• 1	. 1	1	~
5.	Cardobaindia sp.	I	۱	ı	1	ı	ť	1		,		• •	1	١	'	,	ı	• •	•
6.	Caudites sp. A	ı	1	ı	ı	ı	• •	,	• •	ı	•	'	•	,	1	ı	1	2	ı
7.	C. sp. B	t	ı	ı	ı	,	,	1	ı	1		•	1	١	ı	ı	I	1 2	2
*	Cletocythereis sp. A	1	ı	ı	,	ł	ł	,	ı	,	•		7	۱	ı	,	٦	7	2
9.	C. conaliculata	۱	,	ı	۱	,	ı	,	1	i	1	'	' '	۱	I	'	ı	ı	1
10.	C. rastromarginata	ł	ı	1	,	,	,	,	,	1	, ,	'	1	١	t	,	ı	r	ł
.11	Cytherelloidea sp.	ı	ł	ı	1	t	-1	,	,	,	'	•	ł	۱	ı	ı	۱	1	ı
12.	Hemicytherura sp.	ı	1	,	ı	,	ı	ŀ	ı	1	,	•	ı	۱	ı	-	ı	4	ı
13.	Hemicythere sp.	I	ı	~	ı	ı	1	1	1	,	•	•	'	,	'	ı	ı	ı	ι
14.	Hermanites mooneyi	. •	ł	ı	10	4	ъ	1	2	5	, ,	•	•	۱	ł	ı	ı	ı	1
15.	H. parviloha	-:	7	6	9	ı	,	1	21	ī	,	•	ł	ı	7	٦	,	7	4
16.	H. transoceanica	4	-1	•?	80	12	10	Ē	9	1	ص	1		1	ı	ı	7	6	ŝ
17.	Jugosocythereis sp.	Ŷ	1	4	16	4	~	~	33 2	0	, ,	•	1	ł	ł	ı	17	6	\$
18.	Keijia demissa	ı	ï	ï	1	ı	,	1	,	ī	' ,	•	1	١	ı	ł	'	2	e
.61	Loxoconcha heronislandensis	ı	ı	ı	ı	ł	ı			ī	'	. 16	m	ł	ŝ	ı	,	ı	8
20.	L. huchinensis	æ	7	6	32	45	 	õ	17	4	1 8	4	'	6	ī	10	9	61	20
21.	L. insulariaensis	ı	ł	ı	ı	۰	1	,	ı		1	1	1	m	1	ı	٣	11	7
22.	L. Labrynthica	ı	ı	1	ı	١	ī	ł	,	1	1	1	I	ı	1	ł	7		1
23.	L. n. sp. A	1	ı	ı	,	۱	ı	1	,		'	•	1	ı	ı	ı	ı	ı	١
24.	Loxoconchella sp. A	ı	ı	7	7	,	ı	r	2			1	!	I	I	ł	ı	1	ı
25.	Ĺ. sp. B	,	ı	ı	ı	۱	1	ı			•	1	1	I	ł		ı	٦	,
59.	L. sp. C	ł	F	ı	ł	,	-	-	1			•	١	'	1	I	r	r	m
	Mrocypriders ap.	ı	ı	۱	,	, ,	ı	r	r	,	'	•	I	ı	ı		,	ı	١
58.	Morkhoventa theonepicua Vermonidae cabulai		•	• :	' ?	r • -	1 2		1 0		•	 -	ı	1	1	~ -	1 0	ıc	۱ c
	neuro mod occurs. Deeul toruthorai a su	• •	1	: '	; I	• 1	<u>-</u>	, -	, ,	יי הי	• •	; '			1	. ,	, ,	• •	• •
	Orionina so.	ı	1	۱	ŧ	ı	• •	.,	ı	,	'	1	I	I	,	14	Ś	ł	۱
32.	Ornatoleberis sp.	6	ı	4	9	ŝ	2	6	Ē		'	1	,	ı	ı	; I	-	ı	۱
33.	Paracytheridea remani		ı	١	1	1	e	ŝ	I		'	1	1	1	ı	1	ı	10	Ē
34.	Ponticocythereis sp.	ı	,	۱	ı	,		1	,	•	'	ı	ı	ı	ı	-	ŧ	ł	ı
35.	Procythereis sp. A	۱	,	١	,	,	ı	ł	ı	, ,	•	T	۲	•	ı	,	١	ı	ł
36.	P. sp. B	t	ı	١	,	,	ı	ı	1	, I	1	1	'	,	ı	ı	ι	r	۱
37.	Pterobairia maddockeae	ı	ī	١	1	ı	1	ı	r	, ,	'	'	1	ł	ı	ı	ı	I	ì
38.	Semicytherura sp.	ι	ı	١	ľ	1	'n	,	1	,	'	1	'	I	ı	ſ	ι	1	ł
39.	Triebelina sertata	t	,	۱	r	ı	1	-	1	` ,		'	۱	I	ı	ı	ł	ı	ı
40.	Other	10		24	43	37	37 2	5 6	7 3	5 36	2	2	24	17	t	6	6	42	59
	TOTAL	35	Ś	69	51 1	10 1:	27 9	8 18	9 8	8 101	. 15	15	30	23	13	43	52	80 1	05
					<u> </u>	ontin	ied on	next	page	~									

APPENDIX 3-1
BOREHOLE OAR-2/2A (continued)

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									SAM	PLE D	EPTH	ftbs	f)						
	DLAGNOSTIC SPECIES	115.1	127.8	134.0	157.45	171.?	188.25	195.3	204.9	212.45	223.9	234.6	240.0	268.45	282.55	289.7	337.05	379.5	TOTAL
-:	Australimossella sp.	1	,	{ ,	,		 		-	-	-	-	6	5	~	~	•	1	25
2.	Buthoceratina su.	ı	ı	1	,	ı	ı	,	-	,	1	ı	•	'	•	,	,	,	-
i	Callistocythere sp. A	ı	·	ł	,	ı	,	,	ī	ı	,	ı	、 、	1	'	'	ı	,	16
4.	C. sp. B	~	1	ı	-	1	ı	,	۲,	5	13	ſ	۱ 	17	'	1	ı	,	38
۲.	Cardohairdia sp.	t	,	,	Ţ	ı	t	,	ı	ı	,	,	`	•	ľ	ı	١	ı	ŝ
9	Caudites sp. A	ı	ı	ı	,	,	ī	,		,	••	1		1		,	ı	ı	ກ
7.	C. sp. B	ï	ī	ı	,	,	,	,	,	,	ı			1	1	-	r	,	10
8.	Cletocythereis sp. A	ı	ı	r	,	ı	c.ł	5	ı	,		٢	-	'	ı	,	,	1	11
9.	C. conaliculata	·	ı	ı	1	ı	1	1	,	•	ı	6	י ה	'		1	ı		7
01	C. rastromarginatu		-	ı	ı	ł	1	r	ç	-	,	ſ	, ,	'	'	,	ı	·	. .
.11	Cytherelloidea sp.	ł	ı	1	,		1	,	,	,	I.	ı	۰ ۰	1	1	,	1	'	. † 1
12.	Hemicytherura sp.	ı	ı	,	ı	ı	,	1	,	ī	٠	ſ	' 	1	ı	,	ı	ı	¢ '
13.	Hemicythere sp.	۱	ı		,	,	ī	,	.1	1	ł	ſ	, ,	ł	ı	,	ł	,	ۍ : ۲
14.	Hermonites mooncyi	ı	ı	ı	,	,	ı	,	ı	ı	ı	ı	, ,	Ś	•	;	ł	1	ा । न ।
15.	H. parviloba	ı	1	ç	۲	ı	,	1	41	2	53	~~1	ı م	ľ	,	-	ı	ı.	117
16.	H. transoceanica	Ś	7	€ €	ı	1	ı	,	-1	-	14	1	0 13	7	ł	æ	ı	-	142
17.	Jugosocythereis sp.	2	7	4	-	,	ı	1	6	1	13	7	9 20	7	ı	10	ı	S	218
18.	Keijia demisea	t	ı	ı	ı	,	ı	,	ı	ī	ł	ſ	, ,	١	,	,	1	ı	ŝ
.61	Lozoconcha heronielandensis	ł	-1	1	,	ı	t	,	,	ı	1	1	י י	ı	ı	ï	ł	•	36
20.	L. hwahinensis	-	18	30	37	ı	~1	r-	32	34	51	ж Э	1 12	51	01	24	ł	27	679
21.	L. insulardaensis	1	•	,	ī	ī	1	ł	13	8	~	2	7 6	~ ·	7	. + 1	ı	- t	23
22.	L. Labrynthica	2	ı	2	ı	ı	1	,	ų	ı	ı	\$	2	\$	~	5	I		С т
23.	L. n. sp. A	,	•	,	ŀ	ı	1		:	ı	,	ſ	1	10	7	¢	ı	7	25
24.	Loxoconchella sp. A	ı	ı	,	1	1	I	r	:	,	,	1	י י	1	ı	1	t	1	יכנ
25.	L. sp. B	ı	ł	,	1	I	ı	,	ı	i.	ı.	r	1	1	•	ı	ı		ru ;
26.	L. sp. C	ı	64	ı	-	ı	ł	,	1	ı	ı	ſ		1	1	,	ı		71
27.	Miocyprideis sp.	,	ı	t	ı	1	1	,	15	77	¢	,	י י	r 4 -	4	; .	1		л. с
28.	Morkhovenia inconspicua	1	ı	61	1	,	,	,	-7	,	ı	r	3	-	ı		ı		17
29.	Neonesidea schulzi	e S	ı	m	15	-		ł	10	- 7		۲٦	י ה	1	,		1	-	197
30.	Occultocythereis sp.	ı	ı	ı	t	,	r	,	ţ	ł	ı.	r	י י	'	ı	J	1	ı	4 ¢
31.	Orionina sp.	۱	ı	ł	t	,	,	4		I		ſ	1	1	ı	,	ı	•	
32.	Ornatoleberis sp.	ı		Ś		1	1	,	6	ı	с л .	ſ		•	'	ı	ı		25
33.	Paracytheridea remoni	ı	ı	ı	-	,	,	ı	4	ı		ı	י י	•	ı	,	•	-	() -
34.	Ponticocythereis sp.	ı	t	,	,	ī	,	,	,	ł	ı	ł	1 1	•	1	'	ı	• •	;
35.	Procythereis sp. A	١	1	1	ı	ı	1	,	13	4	ŝ	ſ	1 64	!	-	1	ı	1	2
36.	P. sp. B	t	ı	1	,	ı	ı	,	,	ł	11	ī	•	'	,	1	1	•	a ′
37.	Pterobaintia maddockere	r	1	ı	ı	r	1	_	t	1	ł	1	' 1	ŧ	'	ţ	ı	ł	:
38.	Semicytherung sp.	ı	,	ī	۱	1	ı	ī		ī	-	c1	41 2	-	-	ł	I	ı	61
39.	Tricheling sentati	۱	•	ŗ	ı	i	1		-	r	1	ı	י ו	' :	'	•		17	~ ~ ~
.0 1	Other	12	-†-	6	18	,	7	2	38	44	1.	5	8	ζζ	5	61	-	77	858
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	TUTAL	11	42	ç,	76		, c	7	30 1		46	-		- 1	ţ	(.o	-	66	

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									SA	MPLE	DEPTH	(ft	bsf)								
	DIAGNOSTIC SPECIES	0.25	14.15	25.75	38.4	49.7	60.2	66.75	72.8	83.7	89.8	100.45	101 4	119.1	125.25	131.95	137.15	146.1	154.25	165.6	173.05
-	Australimoosella sp.	1	'		,		1	۱,			1				1	1	2	7	1	•	
۰. ۲.	Bythoceratina sp.	I	1	ı	,	ı	,	,	,	1			,	1	ı	I	I	ſ	'	,	,
÷	Callistocythere sp. A	ł	۱	I	18	ı	2	2	-1		5	v	' 	۱	ĥ	٣		-	I	1	ı
4.	C. sp. B	ł	,	t	,	,	,	١	,	ī	1	•	•	•	6	•	:	r	•	-	01
<u>۰</u>	Cardohairdia sp.	ı	ı	7	ł		Ŧ		I					'	ł	1	ı	"	t	1	ı
6 .	Caudites sp. A	ł	ı	ı	ı	1	t	۱	ı	I	1		'	'	'	۲	٠	-		-	1
7.	C. sp. B	,	1	ı	ı	ı	ı	ı	ι		, I	,	,	•	-1	•	•	'	ı	ł	1
ŝ	Cletocythereis sp. A	ı	I	ı	ı	ı	ı	1 ۱	2 1	9	œ		-	1	-	1	2	7		-	2
.6	C. consticutata	۱	١	ı	ı	ŧ	ŧ	ı	1	•			، ۱	•	1	۱	ı	٢	ı	I	1
10.	C. rastromarginata	ı	I	ı	ı	ı	ı	ı	1	1	,	•	•	'	١	I	-1	'	۱	t	ı
.11	Cytherelloidea sp.	,	ł	-1	ı	ı	ı	,	,	1	,	•	, ,	1	ı	1	I	ł	ł	I	'
12.	Hemicytherura sp.	r	ı	ı	•	,	,	۱			1		,	•	-1	I	1	ł	ł	-	-
13.	Hemicythere sp.	ı		ı	ı	,	,	,	,			, ,	-	'	'	I	ı	'	١	ı	ł
14.	Hermonites mooneyi	ı	ï	ч	1		ŧ	1	ı	ı		,	•	•	١	I	e	-	I	7	ı
15.	H. parviloba	1	2	7	ŕ	2	17	61	ı	1	,	' '	•	•	~1	-	9	ł	7	I	ı
16.	H. tronsoceanica	12	11	12	6	7	7			0	.1	י ה	ч ч	1	(1	,	m	æ	9	12	. 2
17.	Jugo socythereis sp.	æ	0	-	ł	2	7	18	ł	•	•	,	-	'	e	١	80	'	9	10	11
18.	Keijia demiser	I	I	ı	ı	ı	ī	1	,	t	,		, ,	•	١	ľ	1	-	-		
19.	Loxoconcha heronislandensis	ı	ı	ı	ı	ı	ı	ı	7	8	ب د	.,	2 23	I	ł	'	ı	'	,	r	1
20.	L. huahinensis	30	10	34	6	24	6	6		1	,		.4	•	1	Ś	35	=	24	15	[]
21.	L. insulardaensis	۱	ı	ı	ı	ı	ı	ſ	,	2	1	•	•	•	ŝ	I	24	12	e,	- •	۱ د •
22.	L. Labrynthica	ı	i	ľ	•	١	ı	1	,				, ,	•	r	1	ſ	7	9	~	10
23.	L. n. sp. A	,	,	۱	,	1	,	ı	ı	•			•	•	1	r 1	,	ł	I	ı	ŀ
24.	Loroconchella sp. A	4	ł	ı			ł	ł	,				,	•	•	١	ſ	,	١	ı	
25.	L. sp. B	ı	ı	ı	1	ı		ſ	•	•				1	t	I	•	, .	ı	ı	-
26.	L. sp. C	١	ı	,	ı	,	ı	,			1	,	'	•	ı	ı	'	-	ı	ı	•
27.	Miocyprideis sp.	,	ı	1	ı	ı	•	r	1	•			, ,			I.	ı	r	1	1	i
28.	Morkhovenia inconspicua	1	1	1	1	,	-	,	ı		1	•	, ,	•	t	1	• •	J	1	1	
29.	Neonesidea schulzi	45	32	41	32	12	32	65				, ,	.~		1	-	1	2	12	62	1
30.	occultocythereis sp.	ı	ł	ı	1	1	۱	1	ı			,	•	•	'		ı	ı	ı	•	I I
	Ortoning sp.	1	•			1	ı	,	1.				•	1		7	ı	,			، ،
32.	Ormatoleberie sp.	-	~	đ	-		1	,	9	^		, ,	• :	•	- 1	1	•		- , ,	4 1	ч.
33.	Paracytheridea remani	7	-1	-	ı	ı	,	,	_	•	•	,		ι	-1	ı ·	Q	7	~	^	c
34.	Ponticocythereis sp.	•	ł	'	ţ	۲	ł	,	1	1				•	ı	-	ı	ı	ı	ı	,
35.	Procythereis sp. A	t	t	ı	ı	,	1	,	1	•	•	•	'	•	ı	ł	•	ı	۱	ı	1
36.	P. sp. B	•	1	,	ı	ı	ı	,	1	•		•	'	•	ı	•	1	1	ı	,	,
37.	Pterohairlia maddockeae	ï	ı		ı	ı	1	,	,	•	•	•	'	•	1	I.	•	1	1	ı	
38.	Semicytherura sp.	ı	I	I	-4	1	t	,	,	, ,	•	•	•	t	ı	ł	7			ı	-1
39.	Triebelina sertata	ı	-1	ł	t	-1	r	,	1	•	•		•	•	' :	•	1	' .	1	1	1
40.	Other	72	31	75	8 7	28	53	36 1	8	5 28	• ~		7		=	م	68	1	51	7.5	0.4
		176	2										75		57	11	168	90	1 2 6	117	141
	10146	[/]	74	1/0	771	1 ()	77 67	-	r 0	t		•	:		;	;		2	ì	3	

(continued on next page)

APPENDIX 3-2

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BOREHOLE OOR-17

									SA	IPLE D	EPTH	(fc b;	()							
	DLAGNOSTIC SPECIES	184.25	193.6	200.9	209.3	215.5	226.05	233.35	239.0	201.7	270.1	285.65	292.1	299.15	310.7	320.2	331.2	339.0	367.9	TOTAL
	Australimoosella sp.	•	' '	•	1			-			'	1	'	•	•	,	-	10	0	14
	Bythoceratina sp.	ı	ı	ł	ı	ł	1	1			-	ľ	١	I	ı	ı	4	ı	,	-
÷	Callistocythere sp. A	ŝ	,	,	1	,	ı	ı	,	1	•	'	5	ł	ı		ı	ł	ŀ	47
4.	C. 3D. B	1	ŀ	ı	,	,	1	2		1	1	'	1	-	1	ı	ı	,	ı	25
<u>،</u>	Cardobairdia sp.	1	ı	,	,	1	. 1	.,	· ,	1	,	I	I	. 1	1	,	,	1	,	, v
	Cruditas an A	-	ı	ı	ł	,	ı				-	-		1		. 1				` :
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	Cletocytherete sp. A	ı	ı	~1	¢	~1	-	•		1	1	1	,	ı	ī	ı	ı	1	ī	113
÷.	c. conaliculata	ı	ı	,	,	,	1	- -	'	'	I	'	ı		ı	1	-	7	,	7
<u>.</u>	C. rastromarginata	r1	ŀ			ı	,	-			'		ı	ı	ı	۱	1	ı	ı	7
11.	Cytherelloidea sp.	1	۱	ı	ı	ı	,	1	'	1		,	,	ı	1	1	t	ł	١	7
12.	Hemicytherura sp.	ı	ı	ı	1	-	1	•	'	•	1	'	1	1	•	۲	-	1	ı	Ś
13.	Hemicuthere sp.	,	7	1	1	,	,	1	'	1	'	'	1	ı	,	,	• •	,	ı	1 -1
14.	Hermonitae moneur	,	ı	1	ı	ı	,	1	,	1	1	'	1	ı	1	,	1	,	ı	· c
5	H. rowifobo	-	1	2.2	v	ı	,					c		-	I					, L
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<u>.</u>	H. transoceantca	æ :		ð	7	4	ı			'	-	4		4	ı	•	1	,	-	177
	Jugosocytherets sp.	Ξ	ı	÷	4	(1	ı	ï	'	•	-		61	1		-	2	7	ı	117
18.	Keijia demisea	1	1	ı	,	ı	ī		•	,	ŀ	ı	١	ı	ı	,	ı	ı	ł	-1
19.	Loxoconcha heronislandensis	ĥ	ı	ı	ı	ı	1	1	'	1	I	ı	-	ı	,	۱	ı	ı	ı	60
20.	L. huahinensis	1	ĥ	35	25	14	7 6	- ~	-	1	16	31	12	26	1	ı	ć	۹	-	475
21.	L. insulardaensis	۱	ŧ	ı	ı	-		د		'	S	~	~	(1	ŀ	,	ı	2	1	67
22.	L. Labrynthica	۰	ı	11	٣	,	,		-	ı	'	١	ı	ı	•	ı	,	1	,	95
23.	L. n. sp. A	ı	ī	ı	ı	,	,	,		1	'	'	ı	ı	1	١	ı	ţ	~	-7
24.	Loxoconchella sp. A	ı	ı	ı	ı	ι	,	,	'	ſ	-	ſ	ı	ı	ī	ı	ı	,	,	7
25.	L. sp. B	ı	١	ı	ł	,	,	.t	•	1	٣	ſ	ı	ı	•	ı	ı	ī	ı	s
26.	L. sp. C	ı	ı	ł	ı	,	1	•	•	I	I	Ś	I	۱	•	ı	ı	ł	ı	\$
27.	Miocyprideis sp.	ı	ī	ı	ı	1	,	,	'	I	1	ſ	I	ı	•	1	1	ı	,	,
28.	Morkhovenia inconspicua	1		ı	ı	(1	,	- 7	1	t	1	ł	ĥ	ı	,	•	ı	,	ŀ	13
29.	Reonesidea echulai	-	١	ŝ	2	2	5	6	'	1	Ē	ı	m	1	•	ı	ı	1	ı	336
20.	Occultocythereis sp.	ı	ł	ı	ı	t	1	•	، ا	1	1	ł	۱	ı	ī	t	ı	•	,	,
Э.	Orionina sp.	ı	ł	۱	1	ı	ı	,		'	ı	'	١	ı	•	1	ı	ı	ī	13
32.	Ormatoleberis sp.	•	1	9	1	1	,	~		1	~	-1	1	•	ı	,	,	1	ı	5
3.	Paracutheridea remani	ı	,	,	ı	ı	1		•	I		с•	I	1	r	ı	ı	ı	7	62
34.	Ponticocuthereis so.	i	ł	ı	ı	ı	. 1			1	. 1	r 1	1	1	ı	,	ı	,	ı	
35.	Procuthereis sp. A	1	,	,	ı	,	1	,	1	1	I	I	~	ſ	-	,	-	,	,	~
36.	P. an. B.	1	,	I	,	1	1		,				- 1	4 1	- •	1		•	I	
	Ptomotorindia maddacheao	,	1	,		. 1	. 1					r				,	- 1	. 1	ı	
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	sentcytherura sp.	1	ı	ı	r	,		-	۰	1	ľ	,	1	1 1	ı	;	•	,	-	61
	Irtenel ina seriata	ı	ı	,	ı	t	1	' 1		1	1	r	ŀ	~	,	·	ı	1	1	^
÷	Uther	~		64	12	17	-	~	~1	'	4	[7	17	38	•	•	•	-		113
	TOTAL	40	ç	88	60	4 1	11	-	, e	-	8	Ê	°,	5			1		ar	1111
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BOREHOLE OOR-17 (continued)

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									SAMI	PLE DI	EPTH (1	ft bsf	~							
	DIAGNOSTIC SPECIES	0.0	2.8	11.8	21.1	33.0	40.05	49.7	58.5	66.35	84.15	93.1	104.55	112.9	121.8	130.0	144.5	147.75	151.55	160.05
:	Australimoosella sp.	1		,	5	ı	-	_	2		- 2	1	•	•	 -	1		.	,	.
2.	Bythoceratina sp.	ı	ı	r	1	1	ı	ı	•	,	-	1	I	1	,	,	ı	1	ı	,
÷.	Callistocythere sp. A	7	,	ł		•	1	ı	•	•		'	ı	ł	ı	ı	ı	ı	1	ł
4.	C. sp. B	-	ı	•	1	. †	,	-	-	•	-	-	ł	•	-	t	ŀ	ı	1	,
ŝ	Cardobairdia sp.	ı	ī	,	۱	,	1	,	•	•	'	•	•	,	ł	ŀ	,	1	r	,
6 .	Caudites sp. A	t	ı	~1	ı	-	1		1	•	•	ı	-	ı	ı	ı	-	ı	1	T
7.	C. sp. B	ł	ī	,	,	ı	,	ı	,	, ,	•	I	ł	ı	~	-1	,	1	,	ı
8.	Cletocythereis sp. A	ı	ı	٣		ī	7	e	م	1	1 2	~1	ł	ı		,	ı	,		L
•	C. conaliculata	ı	ł	1	ī	١	7	,	•	•	'	I	I	ł	-1	t	1	1	ı	ı
10.	C. rastromarginatu	,	ł	ı	ı	,	,	ł	, ,	•	,	ı	I	ı	ı	ı	ŀ	,	ı	ı
	Cytherelloidea sp.		ł	1	ı	ī	1		•		•	r	i	ı	ı	-	•	ı	,	,
12.	Hemicytherura sp.	ı	1	ı	1	,	ı	,	' '	•	'	•	1	ł	ı	ı	ı	ŀ	ı	
13.	Hemicythere sp.	ı	1	-1	ı	2	2	÷	' 1			ı	ι	,	ı	,	ł	,	,	,
14.	Hermanites mooneyi	e	ı	1	~	2	,	-	-	' _	•	•	•	•	ы	1	ı	1	ı	ı
15.	H. parviloba	Ś	ı	01	7	7	,	,		-	1 7	1	-1	ı	10	e	ı	2	e	ı
16.	H. transoceanica	8	7	ı	ı	25	ŝ	8	•		~ ~	~	ł	-	11	_	,	-	2	
17.	Jugosocythereis sp.	7	m	٦	7	ç	11	12	4	1	10	-	7	-	6	ı	4	7		-1
18.	Keijia demiesa	١	I	I	1	-	1	1	,	•	-	I	-1	ı	1	ı	•	ı	ı	1
19.	Lozoconcha heronielandeneis	•	١	-	~1	4	8	11	- 9	•	3 4	ſ	2	-	1	1	1	ī	1	,
20.	L. huahinensis	18	6	11	27	25	24	26	۰.	٠ ٣	31	80	m	4	22	2	9	13	ç	2
21.	L. insulardaensis	ŝ		Ś	2	7	7	ç	-	_	7	-	I	1	ç	ı		_	-	1
22.	L. labrynthica	ı	•	ı	9	1	5		1	•	۰ ۲	2	ŀ	ī	7	1		-	ı	,
23.	L. n. sp. A	ı	1	ı	•	ı	1	,	•	, ,	۰	ı	I	,	1	ł	ı	,	ı	
24.	Loxoconchella sp. A	ı	ı	ı	٦	1	ı	ı	•	•	•	ı	1	ı	•	,	ı	ı	ī	
25.	L. sp. 8	ı	,	,	ı	ı	1	1		`	۱ ۱	,	1	ı	,	ı	•	•	,	ŀ
26.	L. sp. C	ŝ	ı	7	-1	14	,	ł	' 1	•	'	ı	ı	ı	ı	,	ı	,	ı	,
27.	Miocyprideie sp.	•		ŀ	1	ı	۰	•1	•	•	1 ·	I	ı	1	•	1	ł	•	,	,
28.	Morkhoveria inconspicua			ı	ı	m	m		•		m	t	7	ı	ł	ı	,			,
29.	Neonestilea schulzt	33	7	æ	5	61	ł	<u> </u>	7 4		11	-4	ı	,	·	ı	ı	m	-	,
e.	Occultocythereis sp.	ı	1	ı	ı		1	1	•		•	,	ł	ł		ı	,	۰.	ı	,
	Orconing sp.	• •	-	, .	ı	ı			•	•		- 0	•	ı	7	1	ı	-4 •		1
	Urratoleoerie sp.	a (ı	-			_,	.	•		· ·	7	I	1			1.	4	-	
;;	Haracyenericea remant	7	1	t	~	~	¢	n	1		7	ı	ł	1	æ	7	-	ı	ł	,
	Ponticocytherets sp.	ı	1	t	1	t			•		•	ı		۰.	•		ı	ł	ı	1.
: :	Procytherets sp. A	ı		ı	-	1	7	-	•	•	۱	ı	-	-	,	-	۰ ،	•	,	-
36.	P. sp. B	١	1	ı	ı	•	~4	7	•	-	•	ı	-	1	,	ı	1	,	•	,
37.	Pterobairlia maddockeae	1	ı	1	•	ı	ı	ı	т 1	•	1	•	I	ı	ı	•	•	ı	ı	1
38.	Semicytherura sp.	ı	-	-		7	-	1	1	-	-	ı	ı	ı	, †	ı	۱	1	1	,
39.	Triebelina sertata	,	•	,	,	1	,	ı	•	'	'	•	ŀ	,	ı	ı	١	,	,	,
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41.	"Ptped" spectmens	ı	ı	1	~		6	2	'	1	5	ŀ	-		13	ł	۱	ı	-	
	TOTAL	188	32	70	93 3	36 1.	1 13	14 4	4 26	37	140	40	26	1 1	33	61	21	37	6	80
					~	conti	, panu	on nex	t page	(1										

APPENDIX 3-3

BOREHOLE 082-4

									SAMPL	.E DEP	TH (ft	()sf)							
	DIAGNOSTIC SPECIES	156.8	169.8	178.6	182.3	186.9	190.3	195.5	205.1	213.9	225.6	250.0	291.4	30 0.5	314.7	327.1	347.1	356.4	TOTAL
		5	5		5						5	5	5		5		5	5	
 -	Australimoosella sp.	,	-		4	,	-	'	1	ı	,	1	,	,		•	. 2	-	21
2.	<i>Bythoceratina</i> sp.	ı	ı	ı	١	,		, ,			•	•	ı	•	,	י ו	•	۱	4
÷.	Callistocythere sp. A	1	,	•	,	5	1	ь г	•	ı	۱	ı	,	•	ı	•	•	•	27
4.	C. sp. B	ı	۱	-1	١	-1	ŀ	•	•	-	,	ı	ı	ı	t	•	-	١	17
·.	Cardobairdia sp.	ı	۱	ı	,	r	1	'	•	I	'	ı	ı	ı	ı	, ,	-1	I	-
. 9	Caudites sp. A	ı	ı	ı	۱	,	1		,	-	'	,	ı	ı	,	•	'	1	ø
7.	C. 3D. B	ı	,	,	,	,	,	'	•	•	ı	,	,	,	1	·	1	1	m
8	Cletocuthereis sp. A		I		1	1		ي. ا	1	ı	,	c i	ı	ı	ı	,		1	35
6.	C complicate	ı	,	ı	۱	,	,	۱ ,	'	1	I	,	,	ı	,	•	•	I	ſ
10.	C. restromarginata	ı	'	1	۱	61		'	•	1	ı	,	ı	,	ı	1	'	r	11
	Cutherelloiden sp.	,	,	ı	1	1			'	I	ı	,	ı	,		1	•	'	9
2	Homicuthonum an	•	,	ı	,	,	,	• •	'	,	,	1	ı	,	1		1	ł	ł
<u>:</u> ::	liencegenerate ap. Homisisthowe en		ı	ı	1	ı		,	1	,	ı	۱	ı	,	,	'	1	1	10
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4	HETMODILLER MOOREUL	ı .	•	ı .	11		,	, .	• •	,	• •	,	ı	•	,				
15.	H. parviloba	-	1	_	1	5		9	~	,		ı	ı	ı	ı	1 1	7	-	8/
16.	H. transoceanica	ı	ī	r	(1)	ı	م	-1	~	1	1	ı	r		,	'	1	١	32
17.	Jugosocythereis sp.	I	-	ب	9	4		2 17	22	2	80	2	e	ł	-1		1	'	158
18.	Keijia demissa	ı	ı	,	ı	,	1	'	1	•	ı	,	،	,	1	•	•	'	5
19.	Lozoconcha heronislandensis	-	2	,	ı	-1	,		'	,	1	ı	ī	,	,	'	7	ı	57
20-	L. hunhindneis	2	7	4	17	9	5	1 27	12	14	4	~1	ı	-		۔ ۱		I	363
21.	L. insulardaensis	. 1	1	-	4			11	1	1	-	-		,	,	'		٦	88
22.	L. Labrunthica	1	T	-	2			~	4	1		• •	•	1	,	,	•	1	64
23.	L. n. sp. A	,	ı	,	,	,	•	'	'	,	'	ı	,	,	1	1	1	ı	1
24.	Coroconchella sp. A	ı	1	ı	ł	,	,	•	1	1	ı	ι	·	,	,		•	I	-1
25.	L. an. B	,	ı	ı	,	,		'	1	1	,	ı	ı	ı	,	•	1	'	,
26.		I	ı	,	_	,	,	'	1	,	,	ι	ı	,	,	•	'	'	23
77	Wineuruideie sp.	ı	ı	,	,	,		'	'	'	,	,	,	,	,	'	1	1	8
	Monthstania inconstinut	,	1	ı	ç	,		'	'	,	,	-	,	ı	,	,	'	,	17
29.	Norvoeiden erhulzi	,	"	_	10	~			4	ć	~	• 1	ı		,	'	2	ı	180
10.	Decul tocuthere is sn.	I	• 1	, 1			,	. 1	1		• •	۱	1	. 1	,	•		I	ı
11.	Orioning sp.	1	,	ı	ı	,	•	- 22	14	ı	,	ţ	1	ı	ı	•	1	1	45
32.	Ormatoleberis so.	ı	ı	•	ı	ı		-+	1	ı	'	ı	,	ı	ı	-	۳	ı	23
11.	Paracutheridea remani	ı	-	1		2		ł	1	1	I	1	ı	4	ļ	•	1	~	39
14	Dontinonuthonois en	1	• •	1	. 1	. ,	•	, , ,	,	•	,	ı	ı	,	1	.,	•	ı	,
	Decembersis on the	,	,	1	,	,	,	1	1	1	ı	ļ	,	,	,	,	'	,	¢
	crochinereco sp. A																1		· ·
	r. sp. b	1	i	,	,	t	1	1.		ı	f	ı	t	•	1			I	
. / ſ	Pterobatrica maddockere	r	ı	,	ı	•		~~	•	1	1	ŧ	,	1	1	, .		'	- 1
38.	Semicytherurs sp.	,	ı	ţ	,	,	,	'	1		ı	-	,	ı	ı		ι	1	-
39.	Triehelina sertata	:	1	,	ı	,	,	'	۲	ı	ı	¢	ı	ı	1	•	•	'	•
41).	()ther	တ	Ś	Ś	25	1	•	7 51	5	·~	Ś	Ś	e4	ı	۲.	~ ~	1	~	747
41.	"Piped" specimens	ī	•	ſ	••	ı		,	ı	ł	t	ι	1	ŧ	T	1	1	1	; ;
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BOREHOLE OB2-4 (continued)

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$ \begin{array}{cccccccccccccccccccccccccccccccccccc$	$ \begin{array}{cccccccccccccccccccccccccccccccccccc$		DIAGNOSTIC SPECIES	7.0	35.0	44.6	57.85	74.3	89.45	102.0	115.05	131.0	139.7	169.35	174.95	182.3	189.25	198.0	207.3		210.4	229.95	232.1 229.95 210.4	239.15 232.1 229.95 210.4
3. Subjectivities apply and the first apply ap	$ \begin{array}{cccccccccccccccccccccccccccccccccccc$	-	Australimoosella sp.	-4	4	-		ı	1		ı		1	١		-	1	r	ı	ı		-	-	1 - ¹
3. Calificatory how ap. A -1 -1 1 - <td< td=""><td>$\begin{array}{cccccccccccccccccccccccccccccccccccc$</td><th>2.</th><th>Bythoceratina sp.</th><td>-</td><td>-</td><td>-</td><td>,</td><td>1</td><td>ı</td><td>ı</td><td>•</td><td>'</td><td>,</td><td>-</td><td>'</td><td>٠</td><td>I</td><td>'</td><td>ı</td><td>1</td><td></td><td>ı</td><td>1 1</td><td>, , ,</td></td<>	$\begin{array}{cccccccccccccccccccccccccccccccccccc$	2.	Bythoceratina sp.	-	-	-	,	1	ı	ı	•	'	,	-	'	٠	I	'	ı	1		ı	1 1	, , ,
4. C. 91. B. C. T. 91. B. C. 20. B	4. C. 99. 8 $Carrotizes 9. A$ 1 2 1 2 1 4 4 6. Carrotizes 9. A 0 1 2 1 2 1 4 4 4 8. Carrotizes 9. A 1 1 2 1 2 1 4 4 4 8. Carrotizes 9. A 1 1 2 1 2 2 1 4 4 4 9. Carrotizes 9. A 1 1 2 2 1 2 2 1 4	÷.	Callistocythere sp. A	١	-	۱	-		ı	1	ī	' 1	- 2	'	I	'	١	ŀ	ı	ı		,	י י	, , ,
5. $Cauditient sig. 5. Cauditient sig. 5. Cauditien sig. 5. Cauditien sig.$	5. $farriohatiratia set.$ 5. f	4.	C. sp. B		7		ı	ı	7	ı		'	'	7	~	7		4	4	1			1	1 1 2
0. Candicas sp. \mathbf{A} 5 0 1 - <td>6. Classifies sp. A 6 1 -</td> <th>\$</th> <th>Cardohairdia sp.</th> <td>1</td> <td>,</td> <td>1</td> <td>I</td> <td>,</td> <td>• •</td> <td>ı</td> <td>• •</td> <td>'</td> <td>•</td> <td>'</td> <td>١</td> <td>'</td> <td>۲</td> <td>1</td> <td>ı</td> <td>,</td> <td></td> <td>١</td> <td>י ו</td> <td>, , ,</td>	6. Classifies sp. A 6 1 -	\$	Cardohairdia sp.	1	,	1	I	,	• •	ı	• •	'	•	'	١	'	۲	1	ı	,		١	י ו	, , ,
7 C, e_{2} B C, e_{2} C C, e_{2}	7. Constitution (2) 9. C. constitution (3) - </td <th>è.</th> <th>Caudites sp. A</th> <td>1</td> <td>9</td> <td>-</td> <td>ı</td> <td>ı</td> <td>ı</td> <td>ı</td> <td>,</td> <td>•</td> <td>'</td> <td>I</td> <td>2</td> <td>۱</td> <td>ł</td> <td>ı</td> <td>ı</td> <td>I</td> <td></td> <td>,</td> <td>1 F</td> <td> 1 1</td>	è.	Caudites sp. A	1	9	-	ı	ı	ı	ı	,	•	'	I	2	۱	ł	ı	ı	I		,	1 F	 1 1
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14. Ponticocythereis sp. - </td <td>4. Ponticocythereis sp. -<th>33.</th><th>Paracytheridea remani</th><td></td><td>22</td><td>Ś</td><td>7</td><td>17</td><td>,</td><td></td><td>ı</td><td></td><td>2 10</td><td>6</td><td>-</td><td>ĥ</td><td>9</td><td>27</td><td>Ś</td><td>ı</td><td>1</td><td></td><td>5</td><td>2 8</td></td>	4. Ponticocythereis sp. - <th>33.</th> <th>Paracytheridea remani</th> <td></td> <td>22</td> <td>Ś</td> <td>7</td> <td>17</td> <td>,</td> <td></td> <td>ı</td> <td></td> <td>2 10</td> <td>6</td> <td>-</td> <td>ĥ</td> <td>9</td> <td>27</td> <td>Ś</td> <td>ı</td> <td>1</td> <td></td> <td>5</td> <td>2 8</td>	33.	Paracytheridea remani		22	Ś	7	17	,		ı		2 10	6	-	ĥ	9	27	Ś	ı	1		5	2 8
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16. P. sp. 8 - 1 -	6. P. sp. 8 - 1 - <td< td=""><th>35.</th><th>Procuthereis sp. A</th><td>1</td><td>,</td><td>•</td><td>1</td><td>ı</td><td>1</td><td>ł</td><td>1</td><td>1</td><td>'</td><td>1</td><td>1</td><td>I</td><td>•</td><td>ı</td><td>ι</td><td>1</td><td></td><td>r</td><td>•</td><td>1 1</td></td<>	35.	Procuthereis sp. A	1	,	•	1	ı	1	ł	1	1	'	1	1	I	•	ı	ι	1		r	•	1 1
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38. Semicytherura sp. - 4 2 - 1 1 - - 4 - 1 2 -	8. Semicytherwars sp 4 2 - 1 1 1 4 - 1 2	37.	Pterobaindia maddockeae		ı	•	1	1	ı	1	1	'	•	'	1	1	٠	1	r	ı			•	, ,
99. Tricheling servata - <td>9. Triebeling service -</td> <th>38.</th> <th>Semicutherung sp.</th> <td>1</td> <td>4</td> <td>2</td> <td>1</td> <td>-</td> <td>-</td> <td>_</td> <td>1</td> <td></td> <td>-1</td> <td>'</td> <td>-</td> <td>2</td> <td>Т</td> <td>1</td> <td>•</td> <td>ı</td> <td></td> <td>,</td> <td>1</td> <td>۔ ۱</td>	9. Triebeling service -	38.	Semicutherung sp.	1	4	2	1	-	-	_	1		-1	'	-	2	Т	1	•	ı		,	1	۔ ۱
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APPENDIX 3-4

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BOREHOLE OP2-18

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35.	Procythereis sp. A	ł	1	t	ı	1	ı	,	1	, ,	1	•	1	ŀ	,	'	ı.	ı	ı		•
36.	P. sp. B	~	1	٢	,	ſ	1	,	1	1	t	•	'	١	ı	۲	•	,	,	ı	C 1
37.	Pterobaintia maddockege	,	ı	ſ	,	ı	ı	,	•	',	ı	•	ł	ł	•	ı	,	1	ı	ı	ı
38.	Semicutherura sp.	"	I	1	ł	,	ı	ı	1	'	l	1	t	I	ı	,	1	1	ſ	•	~
39.	Triebeling sertata	,	ı	٢	1	1	,	1		,	ı	•	1	1	ı	ŧ	1	1	•	ī	4
40.	Other	÷5	,	~	1		15	1	Ś	1	ſ	-1	ł	1	~	-1	,	1	¢1	ī	88
41	"Ptped" spectnens	1	,	. 1	,		t	ı	,	'	ı	'	ſ	1	ı	ł	١	1	ł	ı	ı
																			l		
	TOTAL	170	61	8	-7		55	1	C.	-	ı	· ·	·†	~	æ	5		~	ç	ī	292

APPENDIX 3-5

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APPENDIX 3-6 BOREHOLE OFT-8

				SA	MPLE	DEP	TH (Et b	sf)		
	DIAGNOSTIC SPECIES	0•0	8.75	18.6	27.9	35.1	43.1	48.85	64.0	74.0	TOTAL
1.	Australimoosella sp.	-		-	-	-	-	-	-	_	-
2.	Bythoceratina sp.	-	-	-	-	-	-	-	-	-	-
3.	Callistocythere sp. A	-	-	-	-	-	1	-	1	-	2
4.	C. sp. B	-	-	-	-	-	-	-	-	-	-
5.	Cardohairdia sp.	-	~	-	1	-	-		-	-	1
6.	Caudites sp. A	-	1	-	-	-	-	-		-	1
7.	C. sp. B	-	-	-	-		-	-	-	-	-
8.	Cletocythereis sp. A	-	-	1	-	-	-	-	1	-	2
9.	C. canaliculata	_	-	-		-	-		-	-	-
10.	C. rastromarginata	-	-	-	-	-		-	-	-	-
11.	Cutherelloidea sp.	-	_	-	-	-	-	-	-		-
12.	Hemicytherura sp.	-	-		-	-	_		-	-	
13.	Hemicuthere sp.	-	1	-	-	_	-	_	-	_	ĩ
14	Hermanites mooneui	_	-	_	1	1	_	_	-	_	2
15.	H. parviloha	1	-	-	-	_	_	-	-	-	1
16	H. transcapanica	-	-	_	٦	4	-	_	-	_	7
17	Jugosoguthongis sp	-	1	-	_	_	8	_	4	-	13
18	Vaijia demissa		-	_	_		-	_	-	_	-
10	Lorogonaha hanonislandonsis	_	r	_	_	_	_	_	2	-	1.
20	L huchingnoig	1	4	1	5	2	2	1	0	1	-+
20.	I, inculanda ancio	· _	4	1	,			1	7		20
21. วา	I labrunthiag	_	1	_	_	_	_	-	1		2
22.	5. Labrynthica	_	-	-	-	-	-	-	-	-	-
2.3.	$J_{i} \bullet n \bullet sp \bullet A$	-	-	-	-	-		-		-	-
24.	Loxoconchella sp. A	-	-	-	-	-	-	-	-		-
<u> </u>	L. sp. B	~	-	-		-	-	-	-	-	-
25.	L. sp. C	-	-		-		-	-	-	-	-
2/.	Miocypriaeis sp.	-	-	-	-	-	-	-	-	-	-
28.	Morkhovenia inconspicua	-	-	-	-	-	-	-	-	-	-
29.	Neonesidea schulzi	-	2	-	-	-	I	-	1	-	4
30.	Occultocythere is sp.	-	-	-		-	-	-	-	-	-
31.	Orionina sp.	-	-	-	-	-	-	-	-	-	-
32.	Ornatoleberis sp.	-	1	-	3	-	-	-		-	4
33.	Paracytheridea remani	-	1	-	2	1	1	-	-	-	5
34.	Ponticocythereis sp.	-	-			-	-	-		-	-
35.	Procythereis sp. A		-	-	-	-	-	-		-	
36.	P. sp. B	-	-	-	-	-	-	-	-		-
37.	Pterobairdia maddocksae	-	-	-	-	-		-	-	-	-
38.	Semicytherura sp.	-	-	-	-	-			-	-	-
39.	Triebelina sertata		-	-	-	-	-	-	-	-	
40.	Other	7	7	2	6	4	4	-	10	-	40
41.	"Piped" specimens	1	-	-	-	-	-	-	-	-	1
	ΤΟΤΑΙ.	10	21	4	21	13	18		29		118

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APPENDIX 3-7

BOREHOLE OKT-13

SAMPLE DEPTH (ft bsf)

	DIAGNOSTIC SPECIES	.4	3.5	.4	3.75	0.0	. 65	6.(3.2	0.0)TAL
_		10	18	25	28	36	5	5	68	80	1
1.	Australimoosella sp.	-	1	-	-	-	-	-	_	-	1
2.	Bythoceratina sp.	-	1	-	-	-	-	-	-	-	1
3.	Callistocuthere sp. A	-	-	-	-		-	13	-	3	16
4.	C. sp. B	3	2	-	-	-		-	-	-	5
5.	Cardohairdia sp.	-	-	-	-	-	-	-	-		-
6.	Caudites sp. A	1	-	-	-	-	-	-	-	-	1
7.	C. sp. B	-	-	-	-	-	2	-	-	-	2
8.	Cletocythereis sp. A	2	2	2	-	-	2	-	3	3	14
9.	C. canaliculata	-	-	-	-			-	-	-	
10.	C. rastromarginata	-	-	-	-	-	-	-	-	-	-
11.	Cytherelloidea sp.	-	1	-	-	-	-	-	-	-	i
12.	Hemicytherura sp.	-	-	-	-	-		-	-	-	-
13.	Hemicythere sp.	-		-	-	-	-		1	-	1
14.	Hermanites mooneyi	-	-	4	5	2	-	-	-	-	11
15.	H. parviloha	2	1	3	8	3	1	-	1	-	19
16.	H. transoceanica	3	3	5	6	3	1	5	3	4	33
17.	Jugosocythereis sp.	5	7	1		3	-	12	3	11	42
18.	Keijia demissa	-	1	-	-	-	1	-	-	-	2
19.	Loxoconcha heronislandensis	2	2	_	-	-	-	-	-	-	4
20.	L. huahinensis	11	19	17	19	11	12	3	8	12	112
21.	L. insulardaensis	-	1	-		-	-	2	2	4	9
22.	L. Labrynthica	-	_	4	-	_	-	2	2	3	11
23.	$L \cdot n \cdot sp \cdot A$	-	-		-	-	-	-	-	-	-
24.	Loxoconchella sp. A	-	-	1	3	_	-	-	-		4
25.	L. sp. B	-	-	-	-	-	-	-	-	_	-
26.	5. sp. C	1		-	-	-	-	-		-	1 '
27.	<i>Hiocyprideis</i> sp.	-	-	-			-	-	-	-	-
28.	Morkhovenia inconspicua	-	-		-		-		1	-	1
29.	Neonesider schulzi	1	3	1'	15	8	1	1	2	-	42
30.	Occultocythereis sp.	-	-	-	-	-	-	-	-	-	~
31.	Prionina sp.	-		1		-	-	12	-	1	14
32.	Ornatoleheris sp.	-	1	1	1	5	-		-	-	8
33.	Paracytheridea remani	4	6	5		2	8	1	3	-	29
34.	Ponticocythereis sp.	-	-		-	-	-	_	_	-	-
35.	Procuthereis sp. A	-		_	-	-	-	-	-	-	-
36.	P. sp. B	-	-		-	-	_	-	-		-
37.	Pterohairdia maddorbae	-	-	1	_	-	-	-	_	-	1
38.	Semicutherura sp.	1	4	_	2	_	2	-	7	-	16
39.	Triehelina sertata	_	-	-	_	_	_	_	-	-	
40.	Other	11	21	29	22	13	7	11	7	26	147
41.	"Piped" specimens	_			-	_	-	_	-		-
	TOTAL	47	76	84	81	50	37	62	43	67	547

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CHAPTER 4:

ELECTRON PARAMAGNETIC RESONANCE STUDIES OF SELECTED BOREHOLE SAMPLES AND DEBRIS MATERIAL FROM OAK CRATER

bу

Carol A. Polanskey¹ and Thomas J. Ahrens¹

INTRODUCTION

Electron paramagnetic resonance (EPR) spectrometry was used to measure the peak shock stress experienced by a variety of carbonate samples as a result of the detonation of the OAK nuclear device. The following results are based on EPR spectra from 136 samples taken from six boreholes and 17 debris samples recovered from the crater floor. Shock pressures were determined by comparing the sample spectra with spectra of Enewetak carbonate samples and Solenhofen Limestone that had been shocked to known pressures in the laboratory. Preliminary work on this procedure was developed by Vizgirda and Ahrens (1980) using Enewetak material from CACTUS crater obtained during Project EXPOE. Their work demonstrated a linear relationship between shock pressure and the hyperfine splitting of Mn^{2+} in the calcite component of CACTUS carbonate samples. The current report contains the analysis of the OAK samples and expands upon the previous calibration technique.

EPR ANALYSIS

The EPR spectrum of calcium carbonate, $CaCO_3$, is a result of Mn^{2+} substituting for Ca^{2+} in a single site in the crystal lattice. The theory of Mn^{2+} resonance absorption in single crystal of calcite is described by Hurd and others (1954). The calcite spectrum is dominated by six hyperfine peaks due to the central transitions M = + 1/2, where M is the electronic magnetic quantum number. The hyperfine splitting results from the coupling between electronic and nuclear magnetic moments (Hurd and others, 1954). The spectrum of a powdered sample of single crystal calcite, Iceland spar, is shown in Figure 4-1. The central transitions are labeled along with the forbidden transitions. Of particular interest to this study are the two outer most peak doublets at the lowest and highest field positions.

Sample Preparation and Spectrometer Settings

The carbonate samples were ground into a coarse powder and placed into Wilmad[®] 707SQ fused EPR tubes. The spectra were taken at room temperature with a Varian[®] E-Line Century Series spectrometer. The calcite spectrum is centered near 3,400 Gauss (G), and ranges from approximately 3,150 to 3,650 G.

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CALCITE POWDER SPECTRUM



FIGURE 4+1. -- EPR spectrum of powdered single crystal calcite. The central transitions are due to M = +1/2, $\Delta m = 0$, where M and m are the electronic and nuclear magnetic quantum numbers, respectively. The forbidden transitions occur when $\Delta m = 1$. G = Gauss.

4-2

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The spectrometer was set at a microwave frequency of 9.56 GigaHertz (GHz), modulation amplitude of 3.2 G, time constant of 0.25 seconds (sec), and microwave power of 20 milliwatts (mWatt). A scan time of 8 minutes (min) was used to obtain the full spectrum over a 1000-G scan range; however, highresolution spectra were also recorded for both the extreme lower and higher field components of the spectrum. For these spectra, the magnetic field was swept over 100 G in 4 min. The high resolution spectra provided greater details of the modification of the hyperfine doublets from the shocked samples. As a result, the high-resolution spectra were used in all of the subsequent analyses.

In addition, all spectra were recorded digitally by the spectrometer. Therefore, it was possible to average several scans to improve the signal-tonoise ratio in samples with low signal strength. Signal averaging proved to be extremely useful for the highly shocked samples because there is a definite correlation between decreasing signal strength and increasing shock pressure. Finally, to remove the slope from the spectrum and reduce the line width of the signal, the spectrometer was operated in the second derivative mode. This was accomplished by setting the modulation frequency of the cavity 90 degrees out of phase with the receiver frequency.

SHOCK PRESSURE CALIBRATION OF EPR SPECTRA

Shock-Wave Calibration Experiments

The calibration experiments were a combination of two different data sets. The first set consisted of Enewetak carbonate samples shocked in the laboratory over 10 years ago. The samples were taken from two different depths, 10 ft and 146 ft, from the borehole XRU-3 located outside of CACTUS crater on the eastern side of Enewetak Atoll. These samples and experiments are described in detail by Vizgirda and Ahrens (1980). One reason for reprocessing these samples was to determine if the shock effects observed by Vizgirda and Ahrens (1980) had changed with time. New spectra were taken of each sample and the results confirmed that the effect of shock on the hyperfine splitting had not altered over the time scale of a decade. Secondly, high-resolution spectra were taken from these samples in order to test the pressure-calibration technique; however, these samples were not used for the pressure calibration which is described subsequently in this Chapter.

A series of Solenhofen Limestone samples also were shocked in the laboratory, and these samples became the basis for the pressure calibration. This material was chosen because its EPR spectrum, also due to Mn²⁺ substitution, is orders of agnitude more intense than the Enewetak samples. The Solenhofen is also more chemically homogeneous, although it is still a polycrystalline material. Limestone cores (diameter 0.25 in.) were cut into 0.4-in.-long cylinders and pressed into stainless-steel sample chambers. The sample chambers were sealed in the rear by a stainless-steel plug which was notched to vent any impact generated gases. The sample chamber was then inserted into a large stainless-steel momentum trap and mounted in a 40-mm propellant gun. Projectiles were made of polycarbonate-resin plastic (Lexan) that contained flyer plates of aluminium or Lexan. These impacted the target assembly at velocities between 0.8 and 1.6 km/sec to yield initial shock

pressures of 1.3 to 9.8 Giga Pascals (GPa). Initial shock pressure (rather than final reverberated shock pressure) is reported, because most of the entropy generated by the shock (and hence the shock damage) is associated with the initial shock wave.

Shock pressures were calculated using the projectile velocities and the impedance match technique (Stoffler, 1972). The average bulk density of the limestone samples was 2.61 g/cm³, and the Hugoniot for Solenhofen Limestone samples was taken from Tyburczy and Ahrens (1986). The remaining Hugoniots for Lexan[©] aluminium 2024, and stainless steel 304 are found in Marsh (1980).

Description of Shocked Spectra

Figure 4-2 shows a series of shocked-limestone spectra. The spectra have all been normalized such that the highest peaks (low-resolution spectra) or the highest subpeaks (high-resolution spectra) are of equal height. The shocked-limestone spectra not only reflect in much greater detail the decrease in the hyperfine splitting as a function of increasing pressure, observed previously in the carbonate, but also reveal that the relative signal strength and width of the two subpeaks also varies in a consistent manner with increasing pressure. It is clear from the last two columns in Figure 4-2 that the extreme low-field subpeak in the low-field doublet and the extreme highfield subpeak in the high-field doublet both decrease in relative amplitude and broaden with increasing shock pressure. Because the spectrum of each doublet is the sum of two individual subpeaks, a change in the magnitude or shape of one subpeak can be enough to create the observed decrease in peak separation of the doublet as a whole. In this case, a shift in the actual line position of either subpeak is not required.

The same general trend in peak variation also can be seen in the shockedcarbonate spectra shown in Figure 4-3. Because of the low signal in several samples, these spectra were not uniformly normalized. The specific behavior of the subpeaks in the high-field doublet is less obvious in this series, and the subpeaks are difficult to distinguish at higher shock pressures. The high-field doublet is ultimately lost in the noise (as seen in the sample shocked to 10 GPa). Another factor which complicates the carbonate analysis is that there is also variation between spectra of some of the "unshocked" carbonate samples. Material from two different depths in borehole XRU-3 was used in the calibration experiments of Vizgirda and Ahrens (1980). It appears that the material taken from a depth of 10 ft is not typical of the bulk of the unshocked samples analyzed from Enewetak Atoll. As a result, a systematic difference exists between the spectra taken from these calibration shots and that of the shocked material from 146 ft depth. Both sets of spectra show consistent variation in hyperfine splitting with increasing shock pressure; however, they do differ in the degree to which they are affected by shock deformation.

A second observation, mentioned earlier, is that the amplitude of the entire spectrum tends to decrease with increasing shock pressure. This effect is much more obvious in the Enewetak samples than in the Solenhofen Limestone samples. A loss of signal could be due to a reduction of the Mn^{2+} concentration in the Ca²⁺ lattice sites. The specific mechanism responsible for this reduction has not yet been identified.







(7). High-resolution spectra of the lowest and highest field components are shown in the latter two culumns and centered at 3,160 G and 3,630 G. FIGURE 4-3. -- Comparison of coral spectra shocked in the laboratory. The full spectrum is shown in the first column and centered at 3,400 Gauss respectivelv.

Pressure Calibration by Differencing Spectra

The previous calibration technique of Vizgirda and Ahrens (1980) relied on measuring the separation, in Gauss (G), of the two subpeaks of the highest field component of each spectrum. The hyperfine peak splitting, HPS, was related to shock pressure, P, by the relationship:

$$HPS(G) = -0.60P(GPa) + 13.85$$
 (high field)

Although the decrease in hyperfine splitting is most evident in the high field component, the signal strength of this peak is also the lowest. Therefore, as the signal intensity decreases, the error in measuring hyperfine peak splitting increases. The following technique was developed to incorporate the variations in hyperfine splitting as well as relative peak amplitudes and widths. In addition, the analysis will work equally well for the lowest field component of the spectrum which always has a higher amplitude.

Digital spectra were used to compare each carbonate sample to a pure, single-crystal calcite standard. Both high-resolution spectra from each end of the spectrum were used in the comparison. The digital spectra consisted of 1000 amplitude values evenly spaced over a 100-G field range. Both sample and standard spectra were first normalized by the amplitude of their respective highest subpeaks. The sample spectrum was then translated along the magneticfield axis until the position of its highest subpeak coincided with that of the standard spectrum. Next the absolute value of the difference in amplitude between the two spectra was calculated for each point over the extent of the doublet. Finally, these individual differences were summed to determine a measure of the "likeness" or "unlikeness" of the sample spectrum to the standard. This number shall be referred to as the integrated difference, or ID, of the sample, which is given analytically by the equation:

$$n=400$$
ID = $\int |Y[\text{standard}] - Y[\text{sample}] / 40G$

$$n=n$$

where n_0 is the index of the amplitude array corresponding to a magnetic field value 20 G below that of the highest peak of the standard spectrum; Y[standard](i) and Y[standard](i) are the normalized amplitudes of the standard and sample spectra, respectively, and N is the number of data points that are integrated. In this case, N is 400. The error in ID is determined by performing a similar calculation, where Y[sample](i) are points in the flat taseline signal on either side of the Mn²⁺ peak.

Figure 4-4 illustrates this procedure with examples of two spectra from the limestone calibration experiments. The first frame shows an unshocked Solenhofen Limestone spectrum normalized, translated, and plotted over the standard calcite spectrum. The second frame is a plot of the absolute value of the difference between the amplitudes at each point over a 40-G range in magnetic field. The final two frames demonstrate the same technique using a sample which has been shocked to 9.8 GPa. The error is determined by using the same scheme to calculate the integrated difference along a flat portion of the spectrum. This value gives an estimate of the contribution of noise to the ID over the region containing the signal.



FIGURE 4-4. -- Illustration of the differencing technique showing (A) an overlay of the standard spectra and an unshocked Solenhofen Limestone sample; and (B) a plot of the individual absolute differences at each point along the field. Frames (C) and (D) correspond to (A) and (B) for a limestone sample shocked to 9.8 GigaPascals (GPa).

SOLENHOFEN LIMESTONE



FIGURE 4-5. -- Plot of the summed differences for the low and high field components of the Solenhofen Limestone samples as a function of shock pressure. The ID value is normalized by 40 Gauss (G), the magnetic field range over which the differences were integrated.

	SOLENHO	FEN LIMESTONE		
SHOT NUMBER	P (GPa)	ID Low Field	ID High Field	
	0.0	20.79	47.61	
	5.3	38.12	85.13	
718	6.9	43.53	93.17	
719	9.8	52.99	106.43	
720	2.6	34.03	77.16	
726	1.3	23.44	55.14	
121	1.8	28.42	68.23	
	ENEWETA	AK CARBONATES		-
SHOT	Р	ID	ID	
NUMBER	(GPa)	Low Field	High Field	
394	1.4	37.43	74.68	
395	1.6	37.83	84.80	
396	0.2	33.67	66.51	
397	0.3	33.00	74.79	
405	4.4	73.85	146.09	
442	1.9	78.25	146.22	
443	1.5	68.33	166.97	
445	4.8	73.61	165.60	ļ
446	10.6	75.70	181.03	
448	7.9	70.71	161.79	

TABLE 4-1. Pressure (Giga Pascal) and Integrated Difference (ID) Data for
High Resolution Spectra from Samples Shocked in Laboratory Recovery
Experiments.

The results of these calculations for the limestone calibration experiments are plotted in Figure 4-5. The ID values are plotted against pressure for both the low- and high-field components of the spectrum. To determine the pressure to integrated difference calibration, a line was fit to each data set using linear least squares. The resulting equations are listed below:

P	(GPa)	=	0.290(ID)	-	5.97	(low f	ield)
Ρ	(GPa)	=	0.152(ID)	-	7.59	(high	field)

The correlation coefficients for the fit were 0.966 and 0.943, respectively. Table 4-1 contains a list of the ID results for both limestone and Enewetak sample experiments. The average ID values are given for shots where several samples were analyzed.

Using the calibration curves above, shock pressures were then assigned to the OAK carbonate samples. In general, the Enewetak carbonate samples have a much weaker EPR signal than those from the Solenhofen Limestone. Therefore, it was necessary to adjust the intercept of the calibration curves to compensate for the average ID value of the unshocked Enewetak samples. It follows that this method will then assign negative pressures to some samples, because the previous adjustment was made to accomodate the "average" background noise. To avoid this obviously unphysical result, and because this technique is not extremely sensitive for low-shock damage, all samples with shock pressures calculated to be below 2.0 GPa were classified as unshocked. Similarly, the high-pressure cut-off was chosen to be 15 GPa. This is necessary because: (1) no data exist for very high shock pressures, and (2) the intensity of the Enewetak sample spectrum is low even at 10 GPa. Samples with shock pressures calculated to be above 10 GPa were classified as highly shocked.

In most cases, shock pressures were calculated for each sample using both the low- and high-field components of the spectrum. These values were then averaged to determine the final calculated pressure.

OAK DATA

Borehole Sample Selection

The borehole samples consisted of uncemented sediments and carbonate rock clasts from boreholes OAR-2A, OBZ-4, OCT-5, OET-7, OFT-8, and OPZ-18, the locations of which are shown on Figure 4-6. Samples are referred by the appropriate borehole name succeeded by depth in feet below sea level (bsl).

The carbonate material from Enewetak is extremely inhomogeneous and consists of a mixture of both calcite and aragonite. Because aragonite does not have a detectable EPR spectrum (Low and Zeira, 1972), samples were selected, where possible, for their high-calcite content. For example, those samples containing carbonate grains replaced by solution-deposited calcite crystals were preferred because they would yield stronger EPR signals. Choosing good sample material is important because it provides a consistent base for analysis, and because it guards against mistaking a sample with an inherently poor EPR spectrum as being highly shocked. The difference between the two cases generally can be recognized by visual inspection, although it is more difficult to assess with numerical techniques. As a result, the samples chosen were much less porous than the CACTUS samples used in the earlier shock-wave experiments; consequently, the OAK spectra tended to resemble the Solenhofen Limestone spectra more closely. For each borehole, the majority of samples were taken from depths above the gamma (γ) geologic crater zone, defined by Wardlaw and Henry (1986, p. 3) as that interval of rock and sediment that is fractured and displaced beneath the crater. A more specific description of each sample analyzed is given in Tables 4-2 through 4-7 (located at the end of this Chapter). Detailed descriptions of each borehole are given in Henry, Wardlaw, and others (1986).

Results of Borehole Sample Analysis

The two boreholes drilled almost directly below the position of the explosive device (ground-zero, GZ, and bathymetric-center boreholes OBZ-4 and OPZ-18, respectively) were the most heavily sampled for the EPR study. A very highly shocked layer of uncemented material was found in samples from borehole OPZ-18 between 399.9 ft and 415.9 ft bsl. This layer was distinguished visually by the characteristic greenish color of the muddy carbonate sand. The shocked zone was interrupted at 412.4 ft bsl by a thin zone of lightercolored material. The location and nature of this shocked material coincides with a zone of Holocene sediments described by Wardlaw and Henry (1986) as a possible example of material that has been injected. The present results are consistent with such an hypothesis since this material most likely originated near the pre-shot sea-floor surface. Three other sand samples above this layer, 386.9 ft, 368.5 ft, and 357.2 ft bsl, were moderately shocked to at most 3.2 GPa. The highly shocked samples were located primarily in the geologic crater zone beta-2 (β_2) --the transition sands-- whereas the moderately shocked material came from zone beta-lb (β_{1b}) (Wardlaw and Henry, 1986; and Chapter 6 of current Report). The remaining 24 of the 31 samples from OPZ-18 appear to be unshocked.

Remarkably, not one of the 26 samples from OBZ-4 showed significant shock damage. Three samples from the β_{1b} zone registered only marginally detectable degrees of shock damage. Sufficient samples were analyzed from the transition sands and vicinity to characterize the core: therefore, it appears that OBZ-4 did not share the same history as OPZ-18.

Thick zones of highly shocked material were found in each of the three northeastern-radial transition boreholes OCT-5, OET-7, and OFT-8. The transition sands have not been identified in any of these boreholes; however, the spectra of the shocked material are similar to those from the shocked material in OPZ-18.

Spectra were taken of 25 samples from borehole OCT-5, drilled 658 ft from GZ. The results of six of these samples define a heavily shocked zone at least 25 ft thick, extending from 285.3 ft to 309.9 ft bsl. This region occurs within zone β_{1b} (early stage collapse rubble), and these samples are also primarily uncemented sands. Aside from the highly shocked material in this region, four widely dispersed samples appear to be moderately shocked. However, two of these samples (368.4 and 464.0 ft bsl in OCT-5) are examples of the aforementioned situation where poor signal quality biases a pressure determination. Simple visual analysis of these spectra suggest that both samples are actually unshocked. The elevated pressures calculated for

OAK CRATER BOREHOLES



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FIGURE 4-6. -- Map of OAK crater showing the location of the boreholes sampled for this study. Bathymetric contours given in 5-meter intervals.

these depths are an artifact of the noisiness of the spectra due to the small fraction of calcite in these samples.

Borehole OFT-8 is located 1,129 ft from GZ, just within the excavational crater (Henry, Wardlaw, and others, 1986). This borehole was sampled at 24 depths. In OFT-8, the region of heavily shocked material begins near the top of the β_{1b} zone and extends downward for approximately 27 ft. Included within this zone were seven heavily shocked samples located between 153.6 ft and 180.9 ft bsl. Bordering this region above and below are zones containing moderately shocked material. Two moderately shocked samples, 149.2 ft and 151.3 ft bsl, were taken at the base of the β_{1a} (late-stage collapse rubble) geologic zone, and three samples deeper in the β_{1b} extending to a depth of 195.3 ft bsl.

The next farthest borehole (OET-7) is 1,374 ft from GZ. Based on seismic-reflection, paleontologic, and lithostatigraphic data, this borehole is thought to be located outside of the excavational crater (Henry, Wardlaw, and others, 1986). The majority of the nine samples were from the GAMMA zone; however, all but the uppermost sample were heavily to moderately shocked. Of the highly shocked samples, six out of seven were uncemented sediment samples. The highly shocked zone extended from 118.9 to 147.5 ft bsl, and a moderately shocked, cemented sample was detected a. 173.6 f. bsl.

Borehole OAR-2A, located 4,458 ft from OAK GZ, initially was sampled only as a reference core; however, six of 21 samples from this borehole have been heavily shocked. Two other samples were moderately shocked to pressures of 3.5 and 4.4 GPa. All of the shocked samples were located within the top 39 ft of the core and the most heavily shocked material within the first 24 ft. The proximity of this borehole to the reef may suggest that some highly shocked, fine-grain ejecta was transported from the slope and deposited at the site of OAR-2A.

The combined results from the OAK borehole sample analysis are presented in Figure 4-7. The solid horizontal line in each panel indicates the present sea-floor depth. The depth and thickness of each zone containing highly shocked material (P > 10 GPa) as a function of the distance of the borehole from GZ is shown in a simplified manner in Figure 4-8.

Results from Debris Samples

The debris analyzed consists of 14 samples collected throughout the crater by submersible and three samples collected by scuba divers from roughly a single site. The former samples are a subset of a series of debris samples analyzed by Halley, Ludwig, and others (1986). Figure 4-9 shows the locations where each debris sample was recovered. The range values that will be discussed in a subsequent section were measured from this map. Unfortunately, the debris samples included in this study were all taken from roughly the same distance from GZ. Only one sample (OAK 201) was recovered at a significantly different range.

The results of the debris analysis are plotted in Figure 4-10A and listed in Table 4-8 (located at end of Chapter). The majority of the debris samples were relatively unshocked; however, all of the highly shocked debris was found



DEPTH BELOW SEALEVEL (ft)

4 - 15

ground zero.



indicates the extent of the borehole sampled, and the solid line defines FIGURE 4-8. -- Illustration of the depth and thickness of regions of highly shocked carbonates recovered from each borehole. The dotted line the highly shocked zone.

OAK CRATER DEBRIS SAMPLES



FIGURE 4-9. -- Map of OAK crater showing the debris-sample recovery sites. Bathymetric contours given in 5-meter intervals.



FIGURE 4-10. -- Results of the debris-sample analysis showing (A) shock pressure as a function of range from ground zero, and (B) shock pressure versus estimated pre-event depth below sea level.

at the base of the reef slope. The reef may have blocked some of the highly shocked material from leaving the crater as ejecta, or this material (ejected from the crater) could have been transported craterward back down the reef slope some time after the crater was formed.

In addition to the range measurements, the estimated pre-event initial depth of a limited number of the debris samples was available from strontiumisotope analysis (Halley, Major, and others, 1986; and Ludwig, Halley, and others, 1986) and gross paleontologic and petrographic analysis by B.L. Ristvet in 1981 (see Ristvet, 1986) and corroborated by subsequent analysis by USGS personnel. The pre-event depth below sea level is plotted against shock pressure for these samples in Figure 4-10. Although the pre-event depth estimates are crude in some cases, a strong correlation is evident between shock pressure and depth for this limited data set. This correlation is consistent with the assertion that the pre-event surface material near GZ was the most severely shocked.

DISCUSSION

With a few exceptions, the bulk of the samples analyzed can be split into two catagories, unshocked and very highly shocked. There were relatively few samples which can be assigned to intermediate pressure categories. This suggests that the majority of the shocked material shares a common origin. Fresumably, the material right at or near the surface near GZ received the highest shock pressure from the blast. During the cratering event, some of this material was incorporated in the lining of the transient crater cavity and was then buried almost immediately by the collapse of the crater walls. Subsequent backwash of ejecta and slumping and deformation of the crater would tend to mix this highly shocked material with other rubble and breccia and consequently obscure any stratigraphically discernable rone of highly shocked material.

This hypothesis can be applied to explain the presence of the shocked regions in OPZ-18 and in the transition boreholes. Because slumping and collapse become increasing'y more important toward the rim of the crater, it is not unreasonable that OPZ-18 is the only borehole to have the shocked material preserved in a stratigraphically discernable unit such as the transition sands. The thickness of the region of highly shocked material does remain fairly constant throughout the three transition boreholes (Figure 4-8), although the region is located at consistently shallower depths as the distance between the borehole and GZ increases. This is a further indication that these regions were at one time related.

Late-stage debris slumping and the influence of sedimentation also have contributed to borehole stratigraphy. Post-event slumps from the reef have deposited at least 8 ft of unshocked debris at OET-7, and possibly as much as 17 ft at OFT-8. The location of the shocked debris samples collected from the floor of the crater suggest that highly shocked ejecta may also be deposited from the reef slope, and the shocked upper layer of OAR-2A coul' be the result of accumulated deposition over many years.

ACKNOWLEDGEMENTS

The authors would like to thank the following people: Lt. Col. Robert F. Couch, Jr., for his support of this project; Thomas W. Henry for assistance with sample acquisitions from the boreholes; Robert B. Halley for providing debris samples; Papo Gelle, Mike Long, and Leon Young for preparing the recovery samples for the shock-wave experiments; and Sunney Chan for use of the EPR facilities.

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TABLE 4-2. Results for Borehole OAR-2A Samples. The pressures and accompanying errors are given in Giga Pascal (GPa). Depths are provided in both fect below scafloor (ft bsf) and feet below sealevel (ft bsl). For explanation of carbonate petrographic names used under DESCRIPTION column in this and succeeding tables, see Henry, Wardlaw, and others (1986, p. 83-97).

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DEF (ft bsf)	PTH (ft bsl)	P (GPa)	ERROR (GPa)	DESCRIPTION
0.6	111.1	12.5	2.6	uncemented sand
6.6	117.1	14.5	2.7	uncemented wackestone
12.2	122.7	13.2	4.5	uncemented wackestone
15.8	126.3	14.5	1.0	uncemented wackestone
21.4	131.9	15.0	2.4	uncemented packstone
23.8	134.3	10.9	1.0	uncemented wackestone/packstone
25.8	136.3	3.5	0.6	uncemented packstone below AA/BB bndry
32.2	142.7	0.0	1.0	cemented packstone
38.4	148.9	4.4	3.0	poorly-cemented packstone
57.3	167.8	0.0	0.7	cemented packstone
67.1	177.6	0.0	1.2	poorly-cemented wackestone
74.2	184.7	0.0	1.3	uncemented wackestone
81.6	192.1	0.0	1.9	uncemented wackestone
100.3	210.8	0.0	1.6	coral fragment, Astreopora
129.3	239.8	0.0	0.6	cemented wackestone
155.3	265.8	0.0	1.4	spar-replaced coral
177.4	287.9	0.0	0.7	well-cemented mudstone
196.3	306.8	0.0	0.8	cemented wackestone
241.6	352.1	0.0	1.2	uncemented grainstone
280.0	390.5	0.0	0.9	cemented wackestone burrow
315.6	426.1	0.0	1.1	cemented wackestone

CRATER	DEPTH		Р	ERROR	DESCRIPTION
ZONE	(ft bsf)	(ft bsl)	(GPa)	(GPa)	
α ₁	6.7	205.4	0.0	0.9	mud
α_2	44.0	242.7	0.0	1.0	wackestone
	75.9	274.6	- 0.0	2.3	coarse-grain packstone
β_{1a}	107.5	306.2	0.0	1.0	cemented
	119.1	317.8	2.3	1.0	cemented packstone
	136.0	334.7	0.0	0.6	cemented packstone
	141.8	340.5	0.0	0.5	cemented
	152.1	350.8	0.0	2.0	cemented
	163.3	362.0	0.0	0.8	cemented
β_{1b}	174.8	373.5	0.0	1.2	spar
	190.8	389.5	0.0	0.7	cemented wackestone burrow
	191.0	389.7	2.3	2.6	lithoclast and spar
	193.2	391.9	0.0	0.6	cemented packstone
	196.1	394.8	2.4	4.2	spar-replaced Favia
	199.6	398.3	0.0	1.6	fine grain muddy sand
	199.9	398.6	0.0	0.7	uncemented wackestone
β_2	207.7	406.4	0.0	1.8	cemented wackestone burrow
	210.9	409.6	0.0	0.7	cemented packstone burrow
	216.6	415.3	- 0.0	0.7	recrystallized Tridacna
	217.1	415.8	0.0	1.2	well-cemented tea-brown micrite
	219.4	418.1	0.0	1.1	spar-replaced coral
β_3	222.7	421.4	0.0	0.7	cemented packstone
	233.0	431.7	0.0	0.7	uncemented
	265.1	463.8	0.0	1.5	poorly-cemented
	324.0	522.7	0.0	1.1	cemented burrow
γ	397.7	596.4	0.0	3.2	spar-replaced coral

TABLE 4-3. Results for Borehole OBZ-4 samples. The pressures and accompanying errors are given in Giga Pascal (CPa). Depths are provided in both feet below seafloor (ft bsf) and feet below sealevel (ft bs!).

TABLE 4-4.	Results for E	lorehole O(CT-5 Samples	. The pre	ssures and a	ccompa-
nying	errors are giv	en in Gi ga	Pascal (GPa)	. Depths	are provided	in both
feet b	elow seafloor	ft bsf) and	l feet below se	ealevel (ft	bsl).	

CRATER ZONE	DE (ft bsf)	CPTH (ft bsl)	P (GPa)	ERROR (GPa)	DESCRIPTION
α ₁	0.9	164.6	0.0	1.0	uncemented grainstone
α ₂	9.4	173.1	0.0	1.1	coarse-grain packstone
	13.0	176.7	0.0	1.0	uncemented packstone
	25.0	188.7	0.0	0.9	fall-in (?)
	36.7	200.4	0.0	1.5	cemented wackestone burrow
β_{1a}	45.0	208.7	4.8	0.6	echinoid spine
	56.7	220.4	0.0	0.8	cemented packstone lithoclast
	63.4	227.1	0.0	2.0	rounded cemented burrow
	85.8	249.5	0.0	1.0	cemented packstone
	95.1	258.8	2.6	1.6	Cardium with internal filling
	104.7	268.4	0.0	3.5	spar-replaced Cardium
	115.7	279.4	0.0	0.8	cemented wackestone
	121.6	285.3	13.6	4.2	uncemented med-grained grainstone
	124.3	288.0	13.3	4.7	uncemented coarse-grained grainstene
β_{1b}	131.9	295.6	14.8	3.6	uncemented grainstone
	135.1	298.8	15.0	8.0	cemented grainstone
	140.6	304.3	12.0	6.3	uncemented Halimeda packstone
	146.2	309.9	15.0	7.8	uncemented Halimeda packstone
	153.4	317.1	0.0	0.6	cemented burrow
	163.6	327.3	0.0	0.6	cemented packstone
	174.1	337.8	0.0	0.8	cemented packstone
	192.6	356.3	0.0	1.6	cemented packstone
	204.7	368.4	2.5	0.4	tea-brown cemented rhizolith
γ	237.0	400.7	0.0	0.8	tea-brown cemented packstone
	300.3	464 0	36	3.5	spar-replaced coral

CRATER ZONE	DE (ft bsf)	CPTH (ft bsl)	P (GPa)	ERROR (GPa)	DESCRIPTION
	8.3	115.2	0.0	0.7	pebble-sized lithoclast
α_2	12.0	118.9	13.0	4.5	uncemented grainstone
	17.8	124.7	-15.0	5.5	coral pebble
	21.2	128.1	11.2	5.2	uncemented Halimeda grainstone
	25.9	132.8	15.0	4.8	uncemented packstone-grainstone
γ	30.7	137.6	12.6	2.7	uncemented packstone-grainstone
	35.3	142.2	13.7	4.3	uncemented packstone-grainstone
	40.6	147.5	13.0	5.0	uncemented fine-grain packstone
	66.7	173.6	6.4	2.1	cemented pebble-sized
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TABLE 4-5. Results for Borehole OET-7 Samples. The pressures and accompanying errors are given in Giga Pascal (GPa). Depths are provided in both feet below seafloor (ft bsf) and feet below sealevel (ft bsl).

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CRATER	DE	РТН	Р	ERROR	DESCRIPTION
ZONE	(ft bsf)	(ft bsl)	(GPa)	(GPa)	
<u> </u>	2.7	133.5	0.0	0.6	tea-brown cemented rhizolith
α_2	6.4	137.2	0.0	0.6	tea-brown cemented lithoclast
	13.1	143.9	0.0	0.6	tea-brown cemented packstone
	17.0	147.8	0.0	0.9	cemented packstone
β_{1a}	18.4	149.2	4.2	1.6	cemented matrix within pelecypod
	20.5	151.3	2.0	1.4	partly spar-replaced coral
	22.8	153.6	15.0	4.0	uncemented packstone
	26.0	156.8	14.5	3.7	uncemented grainstone
	30.4	161.2	15.0	4.5	uncemented Halimeda
	35.0	165.8	15.0	5.2	uncemented packstone
	39.8	170.6	15.0	5.3	uncemented packstone
	45.5	176.3	14.8	2.5	partly spar-replaced coral
	50.1	180.9	14.7	1.0	uncemented packstone
β_{1b}	52.3	183.1	0.0	1.2	Cardium with cemented matrix
	52.6	183.4	0.0	0.9	partly spar-replaced coral
	54.4	185.2	0.0	0.6	moderately cemented packstone
	57.0	187.8	2.6	0.7	moderately cemented Halimeda
	61.1	191.9	2.5	0,9	poorly-cemented packstone
	64.5	195.3	8.1	0.8	cemented shell rubble
	64.9	195.7	0.0	1.1	spar-replaced Astreopora
	67.0	197.8	0.0	1.0	mudstone filled cemented burrow
	73.8	204.6	0.0	0.4	moderately cemented packstone
γ	81.2	212.0	0.0	0.5	cemented packstone
	93.5	224.3	0.0	1.4	spar-replaced Porites
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TABLE 4-6. Results for Borehole OFT-8 Samples. The pressures and accompanying errors are given in Giga Pascal (GPa). Depths are provided in both feet below seafloor (ft bsf) and feet below sealevel (ft bsl).

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CRATER	DEPTH		P ERROR		DESCRIPTION
ZONE	(ft bsf)	(ft bsl)	(GPa)	(GPa)	
	3.6	205.5	0.0	1.4	uncemented mudstone
	7.0	208.9	2.5	2.1	uncemented mudstone
α_1	10.0	210.9	0.0	1.0	uncemented mudstone
	23.3	225.2	0.0	1.4	uncemented mudstone
	57.8	259.7	0.0	2.1	uncemented wackestone
	78.2	280 .1	0.0	1.4	uncemented grainstone
β_{1a}	95.9	297.8	0.0	1.0	uncemented grainstone
	117.8	319.7	0.0	1.1	uncemented grainstone
	135.3	337.2	0.0	0.9	uncemented grainstone
_	155.3	357.2	2.2	0.9	uncemented packstone
β_{1b}	166.6	368.5	3.3	1.4	uncemented packstone
	182.6	384.5	0.0	0.8	spar-cemented grainstone
	185.0	386.9	2.2	1.3	uncemented
eta_2	198.0	399.9	14.7	2.5	uncemented
	198.6	400.5	15.0	6.3	green Holocene wackestone mud
	207.0	408.9	13.6	5.4	uncemented
	210.5	412.4	0.0	1.1	cemented packstone burrow
	214.0	415.9	14.1	2.5	uncemented
	217.0	418.9	0.0	0.9	tea-brown cemented packstone
	217.1	419.0	0.0	0.7	cemented wackestone
	217.5	419.4	0.0	1.4	cemented wackestone burrow
	220.4	422.3	0.0	1.3	coarse-grain spar
<i>.</i>	220.5	422.4	0.0	0.8	cemented packstone
β_3	223.5	425.4	0.0	1.2	cemented packstone burrow
1	232.9	434.8	0.0	0.9	poorly-cemented packstone
	236.3	438.2	0.0	3.5	partially spar-replaced coral
	245.4	447.3	0.0	0.6	cemented wackestone
	256.9	458.8	0.0	2.0	spar-replaced coral
	273.8	475.7	0.0	0.7	spar-ceinented packstone burrow
	320.5	522.4	0.0	0.7	spar-filled grstropod
γ	400.5	602.4	0.0	0.6	cemented wackestone

TABLE 4-7. Results for Borehole OPZ-18 Samples. The pressures and accompanying errors are given in Giga Pascal (GPa). Depths are provided in both feet below seafloor (ft bsf) and feet below sealevel (ft bsl).

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SAMPLE	RADIUS (ft)	ERROR (ft)	P (GPa)	ERROR (GPa)	SOURCE- DEPTH (ft bsl)
78	1053	13	0.0	0.9	-
125a	1273	13	0.0	1.8	200-500
126	1211	13	13.6	4.2	105-140
127	1095	13	3.0	2.0	_
128	1421	13	10.5	8.0	105 - 140
144	1302	13	12.5	2.3	105 - 140
147	1170	30	2.7	0.9	
155	1299	13	0.0	1.4	_
156	1109	13	0.0	1.1	_
156b	1109	13	0.0	0.9	-
158	906	13	0.0	1.0	200-500
166B	1109	13	0.0	1.0	500-700
167B	1276	13	0.0	0.6	_
168A	1155	13	0.0	0.8	_
168C	1122	13	0.0	1.4	300-500
201	2310	16	0.0	1.4	420
1-1	1358	157	13.1	1.6	105-140
1-2	1358	157	15.0	3.4	>140
1-3	1358	157	14.6	1.6	105-140

TABLE 4-8.	Results for OAK	Debris Samples.	The pressures and	accompanying
errors	are given in Giga	. Pascal (GPa). S	ource-depths are co	nverted to feet
below	sea level from Lu	dwig and others	(1987) and Ristvet	(1981).

CHAPTER 5:

BATHYMETRIC STUDIES OF OAK CRATER

By

John L. Peterson¹ and Robert W. Henny²

INTRODUCTION

This chapter summarizes recent work done by the Air Force Weapons Laboratory (AFWL) and the New Mexico Engineering Research Institute (NMERI) in a first-order assessment of OAK crater bathymetry (Peterson and Henny, 1987). The starting points for this study were the 1958 pre- and postshot bathymetric maps and a new 1984 bathymetric map of the OAK crater (ALICE reef) area of Enewetak Atoll (fig. 5-1).

Objectives and General Procedures

The primary objectives were to characterize and to quantify changes in bathymetry resulting both from the detonation of the OAK device and from subsequent geologic processes. A secondary objective was to provide a set of working maps at a common scale of 1:2400 for use both by the PEACE Program and future investigations.

The approach was to prepare contour maps by digitizing and reprocessing each of the three bathymetric basemaps and to construct three isopach maps from the contour-map pairs with the aid of a computer. Areas and volumes were computed by contour interval for each of the isopach maps, and planar and cross-sectional features were examined critically on all six maps.

Terminology

The following terms are used in this Chapter. No attempt is made here to correlate the cratering terms with those used in other portions of this Open-File Report; this can only be accomplished after synthesis of the various data sets (see statement 8 of the Conclusions).

<u>Circular crater</u> -- crater region consisting of an inner circular component, as defined by the minus 145-ft closed contour in the postshot contour maps, which is enclosed by an outer-circular component as defined by approximately the minus 50-ft partially closed contour on the same maps.

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FIGURE 5-1. -- Map showing OAK crater region and areas included in Holmes and Narver pre- and postshot surveys (H&N, 1958a and 1958b) and the U.S. Geological Survey map (USGS, 1984). Area in common to all three basemaps shown in stippled pattern. OAK surface ground zero (SGZ) depicted by "X", and apparent crater shown by dotted line.

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<u>Crater wings</u> -- areas primarily within the reef slope and just beyond the boundary of the circular crater.

- Elliptical crater -- crater region consisting of the circular crater and portions of the crater wings as defined by the closed minus 20-ft contour of the isopach maps.
- Apparent crater -- the final observed crater, composed of the elliptical crater and the encompassing areas of subsidence. Note that the apparent crater boundary extends beyond the mapped areas of this study.
- Differential relief (subsequently abbreviated ∆-relief) -- term used in describing the net positive and negative changes in relief of an area with time. This is depicted by isopach maps showing the areas of change of relief derived from comparison of the digitized bathymetric base maps.
- Subsidence -- term used to denote an increase in negative Δ -relief without subscribing to any particular mechanism.

Units used in this Chapter are those of the original works; metric for the 1984 bathymetry, engineering for the remainder.

DATA BASE

H&N Preshot Bathymetric Map

Prior to the OAK nuclear event (29 June 1958), a bathymetric survey of the site (ALICE-reef area) was conducted between 3 and 26 June by Holmes and Narver Engineering Company (H&N) for the U.S. Atomic Energy Commission (AEC) using standard rod, fathometer, and lead-line surveying techniques (fig. 5-1). Datum was 0.5 ft below Approximate Mean Low Water Spring (AMLWS). The survey, tied into the Eniwetok Ivy Grid Coordinate System (H&N, 1952; U.S. Army, 1970), originally was planned to cover a $6,000- \pm 6,000-ft$ square centered on the OAK surface ground zero (SGZ) and aligned with the oceanward edge of ALICE reef. A baseline approximately 6,000-ft long was established along the highest topographic area of the reef flat (specifically, along a line of sand bars midway on the reef flat) with benchmarks (BMs) placed on 300-ft centers.

A standard rod survey was conducted perpendicular to the baseline at each BM oceanward to near the reef edge and lagoonward to approxi ately the minus 5-ft elevation. Each survey line was continued lagoonward to 3,000 ft beyond SGZ using an LCM vessel equipped with a Raytheon Recording Fathometer. Vessel course was controlled by theodolite from each BM and at 300-ft intervals by triangulation from the two terminal baseline BMs. Vertical control was provided at these 300-ft intervals by a lead-line sounding. No cross-tie survey lines were run, and a data gap of a few hundred feet at the lagoonward reef edge resulted because of the inability of the LCM-vessel to obtain fathometer readings in water shallower than 10 to 15 ft (H&N, 1958a). The resultant bathymetric map was hand-contoured with 5-ft intervals (1-ft intervals above the minus 5-ft contour) (pl. 5-1)¹. The map did not include the surveyed ocean side of the reef for reasons discussed below.

H&N Postshot Bathymetric Map

A postshot survey of the OAK crater area, using the same techniques described above, was run between 16 August and 4 September 1958 (D + 47 to 67 days). Numerous major problems were encountered in relocating the baseline along the reef flat opposite the crater because it was significantly disturbed (lowered and covered with debris) by the event. Eventually, the terminal BMs were located and the baseline reconstructed. As in the preshot survey, there was a data gap between rod and fathometer surveys that probably was increased by the difficult postshot conditions. These conditions also resulted in little of the reef oceanward of the baseline being resurveyed. Finally, toward the end of the survey, operationally imposed time constraints may have resulted in only every other line being surveyed in the far eastern quadrant of the grid. The resultant map (H&N, 1958b), contoured at a 5-ft interval. covered the same area as the preshot survey except oceanward of the reef baseline (fig. 5-1), thus giving a common pre-/postshot map area of approximately 6,000 x 5,000 ft or 30 million sqft (pl. 5-2).

Detailed documentation of the H&N surveys does not exist. Most of the information presented here is from B.L. Ristvet (oral communication, 1986) who has reviewed the original field survey books and maps referenced and has conducted extended discussions with several of the original workers.

USGS 1984 Bathymetric Map

The third basemap used in this study was the bathymetric map of the OAK crater and surrounding area prepared from a detailed echo-sounder survey conducted in 1984 (D + 26 years) by the U.S. Geologic Survey (USGS) during the Marine Phase of the PEACE Program (USGS, 1984; see Folger, Hampson, and others, 1986, for details of the survey). This survey also was tied into the Eniwetok Ivy Grid; however, datum was Mean Lower Low Water (MLLW), which is 0.18 m (0.6 ft) below the MLWS established for the earlier H&N surveys. Most of the echo-sounder data were collected along 25-m- (82-ft-) spaced lines oriented parallel to the reef. Perpendicular tie lines were run on the average at 180-m (590-ft) intervals (fig. 5-2). Thus, the USGS survey had a sampling density greater than four times that of the H&N surveys. Although smaller boats provided some data at shallower water depths, nearly all data contoured were obtained from the 41-m R/V Egabrag II, which, because of her draft, effectively excluded data above minus 4 m (minus 13 ft). Therefore, although the greatly increased sample density allowed a 1-m contour interval and the survey extended a 1,000 ft both farther out into the lagoon and along the reef slope (figs. 5-1 and 5-2), no bathymetric data were obtained from

¹ Plates 5-1, 5-2, and 5-3 are digitized, reprocessed versions of the

referenced original bathymetric maps; these are located at the end of the Open-File Report in the map pocket.



FIGURE 5-2. -- Fathometer lines used in the 1984 USGS study (from Folger, Hampson, and others, 1986, fig 2, p. A-3).

near the reef or on the reef flat itself. This reduced the contoured area common to all three maps to approximately 25.5 million sqft or 85 percent of the digitized H&N maps (see fig. 5-1).

DATA PROCESSING

Digitized Base Contour Maps

All data input and processing were performed using an Arc/Info Geographic Information System software package (ESRI, 1986). Processing was done on a VAX 11/750 computer at the Technology Application Center (TAC), University of New Mexico.

The data-input process for the two H&N maps (pls. 5-1 and 5-2, located in the map pocket at the end of the Report) was complicated because the maps were not on base-stable media. Both were digitized manually from 1:2400-scalebluelines using a 36- x 48-in. Summagraphics Digitizer Tablet operating in the continuous-string sampling mode. All data entered into the system were initialized and recorded in the Eniwetok Ivy Grid Coordinate System.

Digitization of the 1984 USGS basemap (pl. 5-3) required that a photographic enlargement be made from the original mylar map (1:6000). The enlargement was redrafted to separate contour lines along steep slopes within the study area. This redrafted map was photographically enlarged again to increase digitizing accuracy of the contours.

Three minor corrections were required to standardize and update the 1968 map. The first was a simple conversion of metric contours to feet. However, since no interpolation was applied to the converted metric units, non-integer engineering-unit contours were generated. The second was a depth correction. This resulted from the comparison of the water-depth values interpolated from the USGS bathymetric map to those measured at each borchole site during the Drilling Phase of the PEACE Program. Linear fits to these data pairs showed that fathometer depths exceeded borehole-site depths by 1 percent down to minus 120 ft, and that borehole-site depths exceeded fathometer depths by 2 percent below minus 120 ft (E.L. Tremba, oral communication, 1986). Third, only those portions of the USGS map that overlayed the H&N map boundaries were digitized.

All digitized basemaps were quality-control checked by interactive z om editing with a 13-in. Techtronix 4107A Color Graphics Terminal. The maps were scale-corrected by the computer to be compatible for overlaying data sets.

Derived Isopach Maps

Three pairs of isopach maps were computer-generated by digitally subtracting combinations of the three contour maps. The contour-map combinations and descriptions of the resulting three pairs of isopach maps are listed below. The first isopach map of each pair presents negative ?-relief; the second map shows the positive &-relief. All (as plates) are located in the pocket in the back of this Report.

- <u>H&N Postshot H&N Preshot Map Pair</u>: Plates 5-4 and 5-5 display distribution of short-term elevation changes (event to D + 67 days) primarily due to cratering effects.
- <u>USGS 1984 H&N Preshot Map Pair</u>: Plates 5-6 and 5-7 display distribution of long-term elevation changes (event to D + 26 years) primarily due to cratering effects and redistribution of crater-produced and natural debris.
- <u>USGS 1984 H&N Postshot Map Pair</u>: Plates 5-8 and 5-9 display distribution of post-crater long-term elevation changes (from D + 67 days to D + 26 years) primarily due to continued subsidence and redistribution of crater-produced and natural debris.

Figure 5-3 illustrates how vertical differences in elevation were calculated within the Arc/Info computer framework. As indicated, a polygon is formed where the two sets of elevation contours, one from each map, intersect. To account for as much of the elevation variance inside the polygon as possible, the mean between the two contours was always used. The vertical difference for each polygon, therefore, is the difference between the means of the two contour sets. In the production of the isopach maps, vertical differences were computed for all polygons formed by the intersection of one map overlaid on another. Typically, five thousand polygons were formed per isopach map. Areas for each polygon were computed in square feet and stored as associated attributes. The vertical-difference files and individual polygons were then sorted into 5-ft increments by a decision rule that grouped polygons with similar vertical differences (i.e.; 0 to 5 ft, 5 to 10 ft, etc.) into the same file.

To reduce required computer memory for the graphic displays, a dissolve module was run on the computer map files that combined adjacent polygons having the same 5-ft increment. Tabular data used to compute areas and volumes for each polygon were saved separately. The final groupings of polygons, representing an increment of 5 ft of positive or negative elevation difference (Δ -relief) between two maps, was then assigned a color and/or symbol for the slides or a shade and/or symbol for the hardcopy maps presented in this Report (pls. 5-4 through 5-9). The jagged appearance of many boundaries on the isopach maps results from the oblique angles formed by intersecting contour sets.

Map Products

Table 5-1 summarizes all maps produced during this study. Each map was produced as a color 35-mm slide, and selected maps were output as hardcopy at a scale of 1:2400 using a 36-in. Versatec Electrostatic Plotter. Because of the large number of contour increments required to fully delineate the crater and disturbed region, the color slides provided the best means to make firstorder assessments of the maps. Hardcopy maps were necessary for more detailed analysis and publication. The three digitized contour maps are presented as Plates 5-1 through 5-3, and a positive and a negative display for each of the three derived isopach-map pairs are presented as Plates 5-4 thru 5-9, located in the map pocket of this Report.

EXAMPLE OF VERTICAL DIFFERENCE CALCULATION



VOLUME OF POLYGON = AREA OF POLYGON X VERT DIFF

FIGURE 5-3. -- Diagram showing isopach computational grid.

TABLE 5-1. -- Summary of digitized bathymetric map products for OAK crater for PEACE Program. Note that the 10-ft contour increment is depicted on the negative Δ-relief isopach maps (i.e., pls. 5-4, 5-6, and 5-8) for depth increments greater than minus 20 ft.

MAP PRODUCTS

MAP	PLATE	MA	P	CONTO)UR (ft	INTE	RVAL	SLIDE	PAPER	AREA & VOLUME
TYPE		TIT	LE	3.3	5	10	25	SETS	MAPS	SUMMAP:
<u> </u>	5-1	Hãn Pr	eshot	-	+	_	-	+	+	-
Contour	5-2	HEN PO	stshot	-	+	-	-	+	+	-
MAP TYPE Contour Overlaid Contour	5-3	USGS Po	stshot	+	-	-	-	•	+	-
		HEN Post- on HE (contour	N Pre- Overlay s only)	-	+	-	-	+	-	
MAP TYPE Contour Overlaid Contour	-	USGS on HEN I (contour	Pre- Overlay s only)	-	+	-	-	+	-	-
	-	USGS on HEN P (contour	o st- Overla y s only)	•	+	-	-	•	-	-
	5-4	Hén Isopach (Pre-& Post-)	Negative A-relief		+			+	+	+
	5-5	H&N Isopach (Pre-& Post-)	Positive ∆-relief	-	+	-	-	+	٠	+
	-	HóN Isopach (Pre-& Post-)	Combined Po s. & N eg. A-relief	-	-	-	+	•	+	+
Isopach	5-6	USGS/Pre- H&N Isopach	Neg a tive ∆-relief	-	+	+	-	•	•	+
	5-7	USGS/Pre- H&N Isopach	Positive ∆-relief	-	+	-	-	•	•	+
	5-8	USGS/Post-H&N Isopach	Negative ∆-relief	-	٠	+	-	*	•	+
	5-9	USGS/Post-H&N Isopach	Positive ∆-relief	-	٠	-	-	•	+	+

Plus (+) symbol indicates presence of product, minus (-) absence.

ANALYSIS

On comparing the three bathymetric basemaps discussed above (see fig. 5-1) and knowledge of the extent of the apparent crater of OAK, it is obvious that neither the 1958 H&N maps nor the 1984 USGS map continue outward far enough in any direction to fully cover the total area affected by the OAK event. This forms a significant limitation to any bathymetric analysis.

Map Derived Quantitics

Several problems are associated with obtaining numerical values from the contour and isopach maps. These are complexly related to the previously discussed survey-sampling differences and deficiencies. They include the following: (1) the differences in areas mapped between surveys; (2) problems with positioning of the survey and drilling ships; and (3) the continuing redistribution of debris with time. The interpretation of the results are further hampered by the fact that both the pileup of debris from the crater (positive Δ -relief) and subsidence after the event (negative Δ -relief) occurs over nearly the entire map area yet are inseparable solely from bathwattile data alone. However, even cursory examination of the maps shows clearly recognizable Δ -relief patterns that are easily followed from map to map the with time). Therefore, in general, the larger the area over which measurements are averaged, the higher the confidence of those values. Below are presented selected point (depths), line (cross sections), and area larea and volumes) estimates.

<u>Water Depths</u>. -- Table 5-2 compares water depths at each borehole width during the PEACE Program that are located within the map areas. Bevel level depths are those measured in the field at time of drilling and reports introdu-USGS (Henry, Wardlaw, and others, 1986a), whereas bathymetric water depths are the arithmetic mean of the bounding contours (3.3-ft contour interval). These although the precision of the former are probably to within 0.1 ft, the later could be in error by up to 1.7 ft. Additional errors probably occur due to borehole location uncertainties (\pm 10 ft), which could easily translate in several vertical feet in areas of rough postshot terrain.

Because the USGS bathymetry and drilling programs were completed within a year of each other, the differences in water depths provide a measure of the inaccuracies inherent on the USGS contour map. Fourteen of the boreholes exhibit differences ranging from plus 2.9 to minus 1.7 ft, with a mean of only 0.4 ft and an absolute average of 1.6 ft. The other four boreholes (OGT-5. ODT-6, OLT-14, and OUT-24) exhibit differences exceeding 4 ft (range from plu-4.8 to minus 5.8 ft), have a mean difference of 1.9 ft and an absolute average of 4.9 ft. For OLT-14, there was a problem in locating the position of the borehole (see Henry, Wardlaw, and others, 1986b, p. 390-391). For the other three, no trends are obvious nor is the reason for the larger differences known. These differences do illustrate the problem in relying solely on the bathymetric data to obtain point estimates.

Another important observation is that postshot water depths for borcholes located at roughly equal distances, but on opposite sides of SGZ, are similar

TABLE 5-2. -- Summary of water depths and vertical differences at PEACE
Program borehole locations. Water depths are compared between measured
values at borehole sites in 1985 (Henry, Wardlaw, and others, 1986, p. 60, tbl. 10) and interpolated values from Holmes and Narver preshot and
postshot maps (H&N, 1958a, 1958b) and U.S. Geological Survey postshot map (USGS, 1984), compiled from echo-sounding data from Marine Phase of PEACE
Program. All depths given in ft below sea level (bsl); vertical
differences are given in ft. Note that the location of borehole OLT-14
is questionable (see Henry, Wardlaw, and others, 1986, p. 390-391).

WATER DEPTHS AND VERTICAL DIFFERENCES

	BOREHOLE	H&N	USGS DRILL	USGS MAP	USGS	H&N	H&N POSTSHOT VS
	NUMBER	PRESHOT DEPTH*	LOG DEPTH** (1985)	DEPTH* (1984)	1984-85 DIFF.	POSTSHOT DEPTH*	USGS 1984 DIFF.
							•••••
			PA	RALLEL TO	J REEF		
1	0 RT - 20	67.5	101.4	102	-0.9	87.5	-14.8
2	OQT-19	47.5	117.5	115	2.2	107.5	- 7.8
3	OTG - 23	47.5	164.0	166	-1.6	152.5	-13.1
4	OPZ-18	47.5	201.9	199	2.8	197.5	- 1.6
5	OBZ-4	12.5	198.7	199	-0.4	197.5	- 1.6
6	OCT-5	17.5	163.7	159	4.8	142.5	-16.4
7	0GT - 9	17.5	134.8	136	-0.7	122.5	-13.0
8	0FT - 8	17.5	130.8	129	2.0	117.5	-11.3
9	OET - 7	17.5	106.9	106	1.4	92.5	-13.0
10	ODT-6	17.5	90.1	86	4.0	72.5	-13.6
			PERP	ENDICULAR	TO REEF		
1	OUT - 24	1.5	147.0	142	4.8	127.5	-14.7
2	OBZ-4	12.5	198.7	199	0.4	197.5	- 1.6
3	OPZ-18	47.5	201.9	199	2.8	197.5	- 1.6
4	OKT-13	102.5	164.7	166	-0.9	152.5	-13.1
5	OIT-11	122.5	155.0	152	-2.8	147.5	- 4.7
6	OHT - 10	122.5	137.3	139	-1 .5	122.5	-16.3
7	OJT-12	112.5	143.8	146	-1.7	132.5	-13.0
8	ONT - 16	132.5	135.1	132	2.9	122.5	- 9.7
9	OMT - 15	142.5	110.9	112	-1.1	127.5	15.5
10	0LT-14	127.5	139.7	146	-5.8	132.5	-13.0

* From Arc/Info File.

** From Henry, Wardlaw, and others (1986, p. 60, tbl. 10).

5-11

regardless of differences in the preshot water depths. For example, at roughly 900 ft from SGZ, preshot differences in water depths between OUT-24 cf. the reefward side and OKT-13 on the lagoonward side are 101 ft; postshot differences are only 18 ft. At 1,800 ft from SGZ, ODT-6 and ORT-20 differ be 50 ft preshot compared to only 11 ft postshot. Another pair (OQT-19 and OET-7 at 1,400 ft) exhibit preshot and postshot differences of 30 and 10 ft, respectively. These data suggest that the net cratering effects in both the "coral" media and water were about the same.

Except for OMT-15, which lies along the lagoon radial (southwest transect) of a large debris tongue, all 1984 USGS water depths at borehole locations exceed the 1958 H&N postshot depths by 2 to 16 ft (see tbl. 5-2). Although no other trends are obvious, these values represent the minimum net downward displacement (i.e., downward movement of the surface plus any addition of debris that may have occurred between surveys). At OMT-15, there is a 15-ft decrease in water depth which, if valid, can only be explained by t late-time addition of debris possibly from a neighboring high.

<u>Cross Sections</u>. -- Figure 5-4 presents two composite cross sections through the OAK SGZ parallel to (southwest to northeast) and perpendicular to (northwest to southeast) the trend of the reef. Each profile of the composite was prepared by manually digitizing the respective contour maps. Note that a vertical exaggeration of 10:1 results in slopes accordingly out of proportion. The H&N preshot profiles illustrate that the OAK device was placed above a sharp break in slope of the lagoonward edge of the reef. Comparisons of the H&N pre- and postshot profiles show that a large part of the lagoon side of the crater was originally water, and, therefore, most of the ejecta from that side of the crater was water. Within the circular crater, the flat floor is offset lagoonward from SGZ by 300 ft, and sets of terraces on the reefward side of the crater are evident.

The H&N postshot profile, perpendicular to the reef, crosses the most complex portion of the map near the apex of a large debris mass rising over an ft above the preshot level near the 1,900-ft mark. A slightly smaller debris mass, 500 ft further out, is some 30 ft above the preshot level and appears to have built up against and engulfed a preshot coral knoll. The cross section parallel to the reef shows the break at the boundary of the circular crater and the crater wings. Several distinct terraces within the crater are visible.

Comparisons of the 1984 USGS and 1958 H&N profiles show that the entire region subsided. Maximum downward displacement is concentrated in the mid-tolower depths of the circular crater and out into the lagoon. Significantly less downward displacement has occurred on the wings of the crater, whereas, on the reefward side, material has moved up and in toward SGZ. In assessing these profiles it is important to consider that redistribution of sediments probably resulted in material moving out of and into the plane of the cross sections.

<u>Areas and Volumes</u>. -- Tabulated areas and volumes for each of the three computed respect maps are given in Tables 5-3 thru 5-5, located at the end of



FIGURE 5-4. -- Cross sections through surface ground zero (SGZ) of OAK crater.

the Chapter¹, and summarized in Table 5-6. Volumes were computed in 5-ft increments by multiplying the vertical differences for each polygon by their respective areas and then totaling all of the volumes. Dimensions given are for each map area which, as discussed previously, differ somewhat between maps. The data demonstrate clearly that the entire area subsided an average of 23 ft by D + 67 days and another 12 ft during the next 26 years. As the surface of the crater and surrounding areas dropped, so did the coverage of positive relief, from 27 percent of the area at D + 67 days to only 14 percent of the area after 26 years.

MAP CHARACTERISTICS

The following is a first-order assessment of each contour and isopach map in terms of topographic patterns and characteristics. Because it is difficul: to accurately quantify many of the features discussed, dimensions stated are only approximate.

H&N Preshot Contour Map

The northwest one-third (reefward side) of the 1958 H&N preshot map (pl. 5-1) shows the lagoonward side of the reef flat with sand bars along the upper margin. At the wave-break line, there is a well-defined, nearly linear scarp that is distinctly sharper north of SGZ. Approximately 400 ft reefward from SGZ, the scarp is cut by a 400 x 400-ft embayment. Beyond the scarp, a gently sloping shelf, dipping 1.5 degrees into the lagoon, ranges in width from 1.000 ft south of SGZ to less than 500 ft north of SGZ.

The southeast two-thirds of the H&N map (lagoonward side) comprises the reef slope and the lagoon floor, which contains numerous patch reefs. The reef slope, steepest (up to 15 degrees) north and shallowest (up to 5 degrees) south of SGZ, extends 1,000 ft beyond SGZ. Lagoonward from the foot of the reef slope, the lagoon floor slopes very gently (l degree) toward the lagoon interior. Just south of SGZ is a 75-ft deep, 200-ft wide rawine with a steep. 25-degree headwall. This ravine flattens and widens lagoonward over a distance of about 1,500 ft but retains its identity to at least 2,500 ft as a clear path extending through the patch reefs. On the lagoon floor, numerous patch reefs, roughly aligned in two lineaments parallel to the reef at 1,700 and 2,600 ft lagoonward of SGZ, rise as high as 40 ft above the bottom and range up to several hundred feet in diameter. Their elliptical to triangular shapes on the map are due to the 300-ft H&N-survey spacing. Actually, they are in fact smaller and nearly circular as shown in the 1984 USGS map with its nearly fourfold increase in sampling density.

¹ Tables 5-3 through 5-5, summarizing the data calculated from the computer analysis of the pairs of derivative maps, and Table 5-6, presenting the grand summary of Tables 5-3 through 5-5, are all located at the end of the current Chapter. Inspection of the H&N Preshot Contour Map (pl. 5-1) shows that the device was placed at a position along the Alice Reef marked by a large embayment. In addition, SGZ was located near the beginning of the lagoonward edge of the reef slope and close to the head of the large ravine that cuts into that slope. Although water depth, interpolated from the preshot map for the OAK SGZ, was almost 13 ft at shot time, according to B.L. Ristvet (oral communication, 1986), it was closer to 14 ft due to a 1.4-ft tide.

H&N Postshot Contour Map

The most striking feature of the OAK crater is its symmetry with respect to the geometric center (GC), which is offset nearly 300 ft lagoonward of SGZ. This is shown clearly in the 1958 H&N postshot map (pl. 5-2). All contours from the bottom of the crater up to the minus 145-ft contour, averaging 850 ft from the GC, are closed. The minus 125-ft contour, averaging 1,200 ft from the GC, closes except for a 45-degree sector on the lagoonward side. Furthermore, on the same side at roughly 1,500 ft from GC, the minus 100-ft contour closes to within 120 degrees. Slopes within the crater are much steeper on the reef side with distinct terraces and slump features evident throughout. Contours in the preshot embayment area are noticeably more distorted than at other locations along the reef.

A second major feature shown by the H&N postshot map is the extension of the debris blanket into the lagoon. This blanket is dominated by a 3,000-ftlong tongue of material, 1,500 ft wide at the crater edge and 55 ft thick at the highest point. Actually, the maximum thickness must be at least 75 ft due to an estimated subsidence in that region of at least 20 ft. The debris tongue is cut radially near the middle by a 400-ft wide channel closely aligned with the preshot ravine. This channel, breaching the crater rim at 1,200 ft from SGZ, passes between two topographic highs at 1,500 ft and bifurcates against another topographic high at 2,700 ft from SGZ.

A third major characteristic is the difference in the preshot to postshot topography in the area of the crater wings along the reef slope. North of SGZ the postshot contours virtually overlay the preshot contours, whereas south of SGZ, the contours have changed considerably and most of the reef slope clearly has been modified by the event.

Many of the patch reefs surveyed preshot do not appear on the H&N postshot map. Some were obviously destroyed, others buried by debris; however, many were probably not mapped in the H&N surveys. Resolution of this issue will require a better understanding of the exact survey lines used by H&N. The 1984 USGS Map (pl. 5-3), with its greater sampling density, adds important information regarding these features and probably could be used as a base to rectify the H&N maps.

USGS 1984 Contour Map

The 1984 USGS map (pl. 5-3) depicts many of the same features shown in the 1958 H&N postshot map (pl. 5-2), except that, with its fourfold increase in sampling density, features such as the coral patch reefs, crater terraces, and slump regions are much more sharply defined. After 26 years, the crater

.¥4.

is larger but retains its basic circular appearance; the crater wings have broadened, especially to the southwest. The inner component of the circular crater, still defined by the minus 145-ft contour, has expanded in radius about the GC from 850 to 1,050 ft. Contours are noticeably smoother, and slopes within the crater are steeper, particularly along the reef where at least two distinct scarps are now present. The debris tongue continues to dominate the lagoonward side, and the preshot ravine is still clearly visible as a remnant feature. Folger, Hampson, and others (1986) discuss the features of the 1984 USGS bathymetric map in terms of "physiographic provinces" and compare them to observations from the submersible, scuba-diving, and sidescansonar operations.

H&N Postshot - H&N Preshot Isopach Map

This pair of isopach maps (pls. 5-4 and 5-5) documents the areal distribution of Δ -relief (the net changes in negative and positive elevations), referenced to the preshot datum, resulting from OAK and extending to 67 days after the event.

The most striking feature of the map pair is the nature and distribution of the Δ -relief. Areas of positive Δ -relief, ranging up to 55 ft, cover only 27 percent of the total map of which Δ -relief greater than 5 ft (16 percent of the total map) is restricted to areas lagoonward of the crater. Negative Δ relief dominates all other areas and covers 63 percent of the map, approximately one-half of that is outside the elliptical crater. The remaining 10 percent of the area shows no change in Δ -relief.

Although it is likely that at least some debris from the crater extends over nearly all of the area covered by the H&N postshot map, most of the reef and large regions on the crater wings and beyond are at a lower elevation than preshot. Therefore, this isopach map grossly understates the amount of debris present because of the unknown amount of event-related subsidence which is very difficult to isolate and measure. In fact, the total amount of debris is further understated because a substantial amount of the debris mass, particularly on the lagoon side, was water. Also, a small amount of ejecta impacted beyond the map area. And finally, an unknown amount of the debris mass may have been transported beyond the confines of the map.

A first-cut estimate of the downward displacement can be obtained by viewing the upper corners of the map (north and west of SGZ) that contain the reef flat. Most distant from SGZ, at 3,000 ft from SGZ, there are areas with a maximum of 5 ft of positive Δ -relief. In contrast at 2,000 ft along the same radials, but still beyond the elliptical crater, there are regions of 5 to 10 ft of negative Δ -relief. Because the positive Δ -relief is probably due to debris, and because debris thickness should increase toward the crater, it is concluded that at least 10 to 15 ft of negative Δ -relief is present at the 2,000 ft range. High-explosive craters in wet media typically display such downwardly displaced profiles, although large azimuthal variations often exist.

A second striking feature of this isopach pair is the elliptically shaped crater, defined by the minus 20-ft contour, which is in sharp contrast to the circular crater of the postshot contour maps (see pls. 5-2 and 5-3). This

elliptical crater, composed of the inner and outer components of the circular crater and the crater wings, has a long axis (4,000 ft) parallel to the reef and a short axis (2,800 ft) perpendicular to the reef.

4

Difference contours from the deepest point on the crater floor up to the minus 140-ft contour (400-ft radius) are roughly circular and symmetric about the GC of the crater. Above and up to the minus 110-ft contour (1,000-ft radius), the contour lines are roughly circular, but about the SGZ. Above, the largest rates of increasing difference (narrowest contour bands) occur between the minus 60- and minus 20-ft contours and probably represent a series of scarps surrounding the elliptical crater.

The elliptical shape of the crater is primarily due to the crater wings and to the sloping lagoon floor. This suggests that the crater wings, although controlled by the reef structure, are related to the circular crater. The elliptical crater is notably broken along the southeast by a remnant feature of the ravine and its headwall, previously described for the 1958 H&N preshot map. Finally, beyond the crater wings and predominately to the southwest, the <u>en echelon</u> pattern of difference contours suggests successive slumping parallel to the reef and well out into the lagoon.

USGS 1984 - H&N Preshot Isopach Map

This isopach map set (pls. 5-6 and 5-7) documents the distribution of net positive and negative Δ -relief from the preshot datum to 26 years after the detonation of the OAK device. Generally, the same basic difference patterns and features are displayed as at 67 days (previously discussed isopach map), but with some notable changes.

First, the entire area has subsided further so that now 86 percent of the map area exhibits a negative Δ -relief and only 14 percent exhibits a positive Δ -relief. The reef flat in the upper right corner of the map (north of SGZ) indicates an additional 5 to 10 ft of subsidence since the detonation of OAK. At the bottom of the map, 3,300 ft southeast of SGZ, an additional drop of 5 to 10 ft has occurred since the event. The previous maximum high of 55 ft on the ejecta tongue is now only 40 ft, indicating subsidence and possibly some redistribution of debris. Note that the new small circular highs in the bottom right of the map (east of SGZ) are probably artifacts of the higher density sampling by the USGS and were not detected by the earlier H&N surveys.

Second, the elliptical crater, as defined by the minus 20-ft contour, has expanded parallel to the reef in the crater wings and into the lagoon, but has contracted reefward. The net result is an increase of 500 ft in the long axis to 4,500 ft, but only an increase in the short axis of 100 ft to 2,900 ft.

Third, the difference contours near the crater floor have changed from circular to elliptical. However, the contours above that level (i.e., the minus 160- to minus 100-ft contours) have remained circular, expanded considerably, and shifted toward the reef.

USGS 1984 - H&N Postshot Isopach Map

This pair of isopach maps (pls. 5-8 and 5-9) shows the negative and positive changes in Δ -relief relative to the H&N postshot datum (47 to 67 days after the event) caused by redistribution of debris and long-term subsidence. In general, the entire map area is displaced downward increasing from 5 ft at the map boundaries (3,500 ft from SGZ) to 20 ft over much of the lagoon to 30 ft within the crater. Areas of negative Δ -relief now constitute 89 percent of the map area.

Areas of maximum negative Δ -relief are associated with deeper portions of the crater, the debris tongue, and isolated topographic highs in the lagoon. The concentric patterns of increasing negative Δ -relief vary from 5 to 10 ft at the edge of the circular crater (1,700 ft from SGZ) up to 25 to 30 ft just above the crater floor. This indicates that the circular crater has continued to subside with time. The multiple, repeating circumferential patterns in the elliptical-crater walls and along portions of the reef probably represent <u>en</u> echelon slumping of debris.

Beyond the elliptical crater, the debris tongue exhibits 5 to 15 ft greater negative Δ -relief compared with surrounding areas, even at its maximum extent of 3,300 ft, where a negative change of 20 to 25 ft is measured. Localized areas of 30 to 45 ft of negative Δ -relief occur in the lagoon associated with topographic highs. These areas are complex with converging zones of negative and positive Δ -relief probably representing slumping and redistribution of debris.

Areas of positive Δ -relief are associated with the flanks of several topographic highs, suggesting (as mentioned above) movement of debris downslope. The small positive Δ -relief on the floor of the crater is due to infillin and probably masks a 20-ft plus Δ -relief. The positive lineaments along the reef scarp probably reflect movement of reef blocks and washback of debris into the crater. The narrow positive lineaments bordering the extended crater probably represent movement of debris downslope along the crater rim scarps. The positive circular highs on the middle and lower right side of the map are probably artifacts of the previously mentioned bathymetry sampling density. The positive highs in the lower left portion of the map (south of SGZ) are unexplained.

CONCLUSIONS

Based primarily on analysis of the OAK bathymetric data presented herein, the following conclusions are reached:

- The OAK event produced a circular explosion-type crater with debris distributed outward in all directions, probably continuously, to at least 3,000 ft from SGZ.
- 2. The circular crater, consisting of an inner circular component on the order of 850 ft in radius, probably formed initially by ejection and outward flow of material. This expanded outward by crater-wall collapse, slumping, and inflow of material to form an outer circular component. By D + 67 days, and probably much sooner, the circular crater had grown to a radius of 1,700 ft and a depth of 200 ft.

- 3. A very large tongue of debris, 1,500 ft wide at the crater edge and tapering to 500 ft at 3,000 ft from SGZ, was deposited outward onto the lagoon floor. This is cut by a 400-ft-wide channel that closely tracks the preshot ravine.
- 4. Also by D + 67 days, the entire area out to at least 3,000 ft from SGZ had subsided 5 to 10 ft with crater wings forming and expanding along the reef slope on either side of the circular crater. This resulted in an elliptical crater 4,000 ft parallel to and 2,800 ft perpendicular to the reef.
- 5. Over the next 26 years, the entire area continued to subside. This subsidence ranged from a minimum of 5 to 10 ft at 3,000 ft from SGZ up to 10 to 20 ft just outside the elliptical crater. Even greater subsidence occurred within the circular crater, particularly the lower portions, and out on the debris tongue. The length of the elliptical crater increased 500 ft (from 4,000 to 4,500 ft), but the width increased only 100 ft (from 2,800 to 2,900 ft).
- 6. Also, over this 26-year period, debris within the crater, on the debris tongue, and along the crater walls continued to slump. Elsewhere debris was selectively redistributed.
- 7. In retrospect, preshot topographic features (reef, embayment, ravine, and reef/lagoon slope) had a significant influence on the final size and shape of the crater and on the initial distribution and subsequent reworking of debris.
- 8. Finally, it is believed that a synthesis of the bathymetric data with the drilling, seismic, side-scan sonar, and gravity data will lead to a significant improvement in the quantification of the postshot topography which, in turn, should provide substantial improvement in the understanding of the cratering mechanics of the OAK event.

ACKNOWLEDGEMENTS

We are indebted to Gar Clark and his staff at the Technology Application Center (TAC), University of New Mexico, for their excellent and painstaking work in digitizing the basemaps and producing the contour and isopach maps. We also greatly appreciate the extended discussions with B.L. Ristvet of S-Cubed on the many details of the 1958 H&N surveying and for providing us the last "originals" of those maps.

This effort was jointly funded by the Defense Nuclear Agency (DNA). Project Officer for DNA was Lt. Col. Robert Couch; the Air Force Systems Command Project Officer was Maj. William Clark.

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TABLE 5-3. -- Summary of areas and volumes calculated from derivative map pair formed by combination of Holmes and Narver (H&N) preshot bathymetric/topographic map (pl. 5-1) and H&N postshot isopach map (pl. 5-2). Corresponding maps are Plates 5-4 for the negative Δ-relief and Plate 5-5 for the positive Δ-relief, respectively.

A-RELIEF	CONTOUR	AREA	TOTAL	VOLUME
CATEGORY	INTERVAL (ft)	(sq ft)	MAP AREA (2)	(cu ft)
	50-55	1 34.8	0 004	74.160
	45-50	4,176	0.01	208.796
	40-45	6.201	0.02	279,045
Positive	35-40	21,735	0.07	869,344
	31)-35	122,071	0.4	4,212,440
∆-relief	25-30	299, 152	1.0	8,480,567
	20-25	317,622	1.1	7,940,551
	15-20	727,271	2.4	14,545,419
	10-15	1,265,755	4.2	18,986,328
	5-10	2,118,996	7.0	21,187,252
	>0-5	3,134,338	10.4	13,788,992
Total Posi	tive ∆-relief	8,018,865	26.6	91,132,994
Total Zero	(0) Δ-relief	3,223,946	10.6	0

	>0-5	4,597,830	15.2	19,166,933
	5-10	3,090,431	10.2	29,015,023
	10-15	1,800,139	6.1	27,033,099
	10-20	1,133,271	3.8	22,237,414
	20-21	100,013	2+5	11 477 674
	20-30	4341343	1.5	13,423,374
	30-35	3/3,/10	1.2	12,903,993
	33-40	116 080	1.1	13,017,430
	4(1-4)	2/01909	0.4	12,000,000
	4) -) () 5) - 5 5	256 168	1.0	13,220,082
	55-60	296,106	0.0	1 3 4 7 3 3 3 7 10
	51 - 65	754 941	0.8	14 695 617
	5-20	234,341	0.6	10,407,412
	70-75	210,011	0.9	10,031,370
	75-80	264, 216	0.4	10 407 811
	80-85	274 517	0.0	23 26 7 576
	85-90	269 777	0.9	24,247,570
	90-95	296 963	1.0	27,100,155
	95-100	118 887	1.0	13 6 77 773
Negative	100-105	291.624	1.0	30 298 728
	105-110	346.830	1.1	17.791.053
A-relief	110-115	355.535	1.2	40.539.123
	115-120	320.563	1.1	38,067,852
	120-125	191.098	1.3	48.674.321
	125-130	401.628	1.3	51,943,142
	130-135	324,904	1.1	43.586.633
	135-140	230,240	0.8	31,933,508
	140-145	76,917	0.3	11,138,920
	145-150	61,249	6.2	9,487,313
	150-155	63,632	0.2	9,862,944
	155-160	55,269	0.2	8,843,002
	160-165	41,603	0.1	6,864,133
	165-170	32,152	0.1	5,465,870
	170-175	. 8,008	0.1	4,901,403
	175-180	22,978	0.1	4,135,995
	180-185	2 1,807	0.1	4,404,134
	185-190	7,449	0.03	1,415,289
Total Menn	tive A-calief	10 051 540	() 0	

HEN PRESHOT VS HEN POSTSHOT -- VOLUMES AND AREAS

5-21

TABLE 5-4 Summary of areas and volumes calculated from derivative map	pair
formed by combination of U.S. Geological Survey (USCS, 1984) postshot	
isopach map (pl. 5-3) and Holmes and Narver (H&N) preshot map (pl. 5-1).
Corresponding figures are Plates 5-6 for the negative A-relief and Pla	L+-
5-7 for the positive Δ -relief, respectively	

-

A-RELIEF	CONTOUR Interval (fi)	AREA (øq ft)	TUTAL MAP AREA (2)	VOLUME (cu ft)
	35-40	3 874	0.01	142,848
	30-40	1.5 5674	0.01	1.431.653
o eitive	25-30	81 634	0.30	2,269,071
Galcive	20-25	227 615	0.9	5,202,429
-ralief	15-20	203 841	0.8	3.683.457
	10-15	199.046	1.6	4,865,145
	5-10	897.017	3.5	6,635,921
	0-5	1.678.303	6.5	3, 792, 121
		110101202		26 (22 6/5
fotal Posi	tive A-reliet	3,233,893	13.7	28,022,045
		2 0 0 2 0 0 0		0 309 914
	0-5	2,933,800	11.4	7,000,014 30 190.117
	5-10	3,828,724	13.0	2011001101
	10-15	3,220,200	14.)	40,031,730
	15-20	2,359,641	9.2	40,922,240 76 888 751
	20-25	1,141,411	4+0 2 4	18 651 17
	20-30	676,404	2.0	10, 91, 91, 91
	30~35	4 30,900	1.1	14 686. At
	5)-40	371,007	1	13.924.60
	40-4)	71 7 580	1.	12, 13, 24, 000
	45~50	302,000	1	15 845 (18)
	50-55	304,003	1 7 2	16 (78.99)
	5.7-60	207,000	1.1	17.264.64
	65-70	2/3,902	1.0	17.161.790
	71) = 75	2/5,000	1.0	17,101,17
	70-73	243,220	0.0	17 366 76
	75-80	223,971	1.0	21, 706 35
	80-85	202,303	1.0	20,740,20
	85-90	288,305	1.1	20,004,72
	90-95	286,225	1.1	20,002,01
Negacive	95-100	210,195	1.1	1
	100-105	317,440	1.2	34 34 3 4 5
A-reliei	105-110	338,915	1.3	36,342,01
	110-115	292,983	1.1	33,043,10
	115-120	256,188	1.0	
	120-125	339,398	1.5	41,804,41
	125-130	306,752	1.2	(1 17/ 34
	130-135	329,502	1.3	43,/34,94
	137-140	274,305	1.9	17.010173
	140-145	285,822	1.1	4,580,44
	145-150	387,000	1.5	17,004,01
	150-155	244,093	1.0	37,287.90
	155-160	127,398	0.5	20,112,11
	160-165	70,272	0.3	11,433,36
	165-170	67,390	0.3	11,340,48
	170-175	54,923	0.2	9,519,20
	175-180	48,496	0+2	8,659,31
	180-185	58,401	0.2	10,717,49
	185~190	41,979	0.2	7,869,61

IEN PRESHOT VS USGS POSTSHOT -- AREAS AND VOLUMES

5-20

TABLE 5-5. -- Summary of areas and volumes calculated from derivative map pair formed by combination of U.S. Geological Survey postshot map (USGS, 1984) and Holmes and Narver (H&N) postshot isopach map. Corresponding figures are Plates 5-8 for the negative Δ -relief and Plate 5-9 for the positive Δ -relief, respectively.

∆-RELIEF CATEGORY	CONTOUR INTERVAL (ft)	AREA (sq ft)	TOTAL MAP AREA (%)	VOLUME (cu ft)
	35-40	478	0.002	16,843
	30-35	844	0.003	26,802
Positive	25-30	3,856	0.02	102,241
∆-relief	20-25	17,189	0.1	378,356
	15-20	112,704	0.4	1,976,824
	0-15	272,512	1.1	3,386,556
	5-10	721,865	2.8	5,217.536
	0-5	1.820.125	_7.1	3,739,364
Total Posi	tive ∆-relief	2,949,573	11.5	14,844, 523
	0- 5	4.053.640	15.8	11 485 025
	5-10	6.555.275	25.5	51,868,217
	10-15	7,707,037	30.0	97,957,905
	15-20	3.372.562	13.1	59,101,780
Negative	20-25	857.090	3.3	19,181,484
∆∙relief	25-30	178,053	0.7	4.838.022
	30-35	28,472	0.1	921.606
	35-40	6.822	0.03	257.019
	40-45	3.177	0.01	136,916
	45-50	2,052	0.01	99.545
	50-55	579	0.002	30,640
Total Negat	tive ∆-relief	22,764,759	88.5	245, 8 78,159

H&N PRESHOT VS USGS POSTSHOT -- AREAS AND VOLUMES

TABLE 5-6. -- Grand summary of areas and volumes of negative, zero, and positive Δ-relief for OAK crater area. Summary derived from Tables 5-2 through 5-5. Area given in sq ft, volume in cu ft, net Δ-relief in ft.

τγρε Δ-Β	ELIEF	HAN PRESHO HAN POSTS	DT VS. SHOT	USGS POST VS. HEN PI	rshot Reshot	USGS PO VS. HAN	DSTSHOT POSTSHOT
TYPE Δ-RE AREA: (s Positive Negative Zero (0) VOLUME: Positive Negative AVERAGE N Δ-RELIEF		Value	Percent	Value	Percent	Value	Percent
AREA: (sq ft)						
Positive	∆-relief	8,018,865	26.56	3,533,893	13.73	2,949,573	11.47
Negative Zero (0)	∆-relief ∆-relief	18,951,568 3,223,337	62.77 10.68	22,197,319	86.27	22,764,753	88.53
	TOTAL	30,193,770	sq ft	25,731,212	sq ft	25,714,302	sq ft
VOLUME:	(cu ft)						
Positive Negative	Δ-relief Δ-relief	91,132,994 789,972,178	10.39 89.61	28,022,645 935,058,002	2.91 97.09	14,844,523 245,878,159	5.69 94.31
	NET	(693,809,189)	cu ft	(907,035,357)	cu ft	(231,033,636)) cu ft
AVERAGE 1 Δ-RELIE	NET F: Π	inus 22.99 f	t	minus 35.2	5 ft	minus	8.98 ft

CHAPTER 6:

CONSTRAINTS ON DENSIFICATION AND PIPING FOR THE OAK EVENT

Bу

John G. Trulio¹

BACKGROUND AND SUMMARY

PPG (Pacific Proving Grounds) sites differ widely from typical CONUS (Continental U.S.) sites in structure and composition. Hence, plausibly, high-yield near-surface nuclear explosions might dig much different craters in one setting than the other. But do they? The question cannot be answered by direct comparison of craters from such bursts. It therefore raises the kindred one of mechanism: The crater from a given burst could vary greatly from a CONUS site to the PPG, because dominant cratering mechanisms might -but do they?

A "subsidence hypothesis" proposed in the early 1980's got to the physical nub of this issue: 2

Explosive loading causes widespread fracturing of PPG coral, whose parts then settle slowly under gravity to form the outer one-half to three-fourths (in radius) of the apparent crater -- its "wing." By contrast, the inner one-half to one-fourth grows in several ways, including ejection of solid; indeed, virtually all ejecta come from that inner region -- or (hence) "excavation crater."

In sum, the subsidence hypothesis posits cave-in of a "coral" skeleton³ to fill the space left by water flowing out of it. Here, we call that process "simple subsidence." Its hallmark is an increase in coral density, since coral solids are denser than the water they replace [but for that, gravity (its cause) could not drive it]. Hence, alternatively, we speak of simple subsidence as "densification."

¹ Applied Theory, Inc., Los Angeles, CA 90036.

² The basic idea appears to have been suggested independently by S. Blouin, H.L. Brode, and B.L. Ristvet, years before the PEACE Program began. In the form stated here, the simple subsidence hypothesis is credited mainly to K.D. Pyatt and K. Kreyenhagen.

³ The OAK medium is referred to herein simply as <u>coral</u>. Said medium is a mixture of carbonate sediment, carbonate rock, and sea water with small amounts of other substances (see Chapter 7 of this Report for details of composition of the OAK medium). Used as an adjective herein, the meaning of coral is controlled by the noun it modifies; for example, "coral solid" denotes the solid components of the medium just described.

PEACE Program data do tell of excavation craters about one-fourth to onethird as large in radius as present apparent craters, widened by later slumping of their walls to about 0.4 of the latter radii (B.R. Wardlaw, oral communication, November 9, 1987). Thus, if the wings of apparent PPG craters did form by simple subsidence, then, for a given burst, most CONUS craters would have half the radii (or less) seen at the PPG. By the same token, coral under the wings would be denser now than pre-shot. PEACE Program measurements [borehole gravimetry and gamma-gamma (γ - γ) logging], however, disclose only minor changes in density there: -Layers of coral (roughly horizontal) from the sea floor to clearly identified interfaces below have thinned much more than the measured densities alone imply. Hence, on the available data, most of the sea-floor lowering had other causes than simple subsidence. Succeeding sections summarize the evidence for and against this last statement; though not airtight, the case for it is strong.

When the mean density of solids in a column grows by a smaller factor than the column's vertical compression, lateral transport must take place. Such transport can occur during plastic flow, as in a tube of toothpaste. Another kind, termed "piping", calls for the flow of slurry (here, water plus coral particles) to the sea floor, where currents may sweep it out of the crater. Signs of piping abound in the OAK crater (Wardlaw and Henry, 1986b, p. 10; Halley and others, 1986, p. 4), but not in its wing, reducing the importance of PEACE measurements as constraints on piping (nonetheless discussed below). Plastic flow, perhaps with some "internal piping" (transport), seems the most likely means whereby the wings of OAK's crater formed. If so, similar wings could form at most CONUS sites -- and early, relative to such gravity-driven processes as slumping and densification. For structures, the wing would still be more benign than the excavation crater, but operating there would be no cinch.

BASIC FACTS AND PARAMETERS

Both the OAK and KOA craters were explored during the PEACE Program, but emphasis fell on OAK because many nearby shots preceded KOA; crateringmechanism puzzles are made knottier by the effects of prior shots (example: How did MIKE affect KOA coral?). Indeed, even with the focus on OAK, and OAK's relative simplicity, the data base for assessing density changes remains slim. Priority rightly went to OAK.

In the OAK crater area, vital maps of the sea floor were drawn before the event, shortly after, and during the PEACE Program (see Chapter 5 of this Report). The bathymetric maps tell us how far the sea floor has sunk as a result of the shot. That does more than quantify what it is that we have to explain (essential enough). For, PEACE exploration has shown that, in the wing and beyond, the coral is split into layers by clearly identifiable Lagrangian surfaces (termed "horizons" by the geologists) that are critical here; several appear in Figure 6-1.

The horizons' great value lies in knowledge of their undisturbed (hence pre-shot) depths. By geologic means, those depths are reproducible down boreholes to within a few tens of feet, and most often to ± 10 ft, in this part of the atoll. More important still, they can be located generally to within a foot in any one borehole. Thus, the depth of each has been determined in boreholes inside the crater and out. So, therefore, has the shortening of



1

FIGURE 6-1. -- Vertical sections at the OAK site. Contours C and D, respectively, mark the bases of regions of (a) measurable decrease in sonic wavespeed and (b) measurable downward displacement. The SCALE (in ft) applies to all distances.

OCEAN ----

1000 ft

SCALE

с

- LAGOON

vertical columns between the sea floor and these horizons.¹ Mapped out too is the base of the region of sensible downward displacement of coral below the crater (contour D; fig. 6-1); its border lies close to the contour (C; fig. 6-1) that marks the limit of the region in which seismic wave-speed has decreased measurably (see Chapter 7 of this Report, particularly tbls. 7-2 and 7-4). Evidently, the shortening of any vertical coral column, flagged by lowering of the sea floor, takes place above the contour D -- and the column's mean vertical shrinkage is given by the ratio of sea-floor lowering at the top of the column, to the column's pre-shot height (current height plus sea-floor drop).

The OAK boreholes lay along two lines through the center of the crater (see Chapter 1, fig. 1-3), one parallel to the reef ("reef-wise") and the other at right angles to it ("cross-reef"). By design, most density logs (and results on densification) came from the reef-wise line (see Chapter 2, fig. 2-2). The reason: We need pre-shot density profiles to compute density changes and their effects. Now -- long after OAK -- those profiles have to come from logging in near-pristine coral outside the crater. For geologic reasons, however, systematic changes in material properties occur along crossreef lines; on well-chosen reef-wise lines, the medium is subject mainly to smaller, random, local variations. Most actual borehole locations were chosen for OAK by PEACE geologists on that basis. The line those holes form runs close to the crater's center, so that density profiles along it can stand as rough cylindrical crater-averages at their respective reef-wise stations. What "rough" means rests with actual PPG measurements; so does the gut issue of reef-wise fluctuations in natural density profiles (below).

Table 6-1 presents basic PEACE data from the OAK crater, including the mean compressions of coral columns at borehole locations. All OAK boreholes are listed in the table; for reference, so is the estimated peak airblast pressure that acted above each. Of special weight are the table's mean vertical shrinkages $\Delta D/L$. Their average of 13 percent (16-1/2 percent for the reef-wise holes), if achieved via simple subsidence, would entail a mean density increase of ~.13 g/cc (.18 g/cc reef-wise). That is larger by almost a factor of ten than the limit of BHG (and γ - γ) resolution achieved in the PEACE Program (.01 to .02 g/cc). Hence, direct on-site evaluation of the subsidence hypothesis was indeed feasible (not known when the program began). Concern therefore lies instead with systematic error and the natural reef-wise scatter of density/depth profiles. Further, as readily confirmed,

¹ The pre-shot depth of the sea floor is known at borehole locations, as are the depths of some horizons (depth uncertainties and confidence questions are taken up later). For a given borehole and horizon, the difference between horizon depth and sea-floor depth is the pre-shot height of the vertical column between horizon and sea floor. Likewise, for that same borehole and horizon, the post-shot height of the column from horizon to sea floor is also known. Between those two levels (horizon and sea floor), the particles of solid in the column may or may not be the same pre-shot as post-shot. The simple subsidence hypothesis says they are the same. The hypothesis is tested herein by adopting it, and comparing the column shortening it implies (when coupled with measured densities) to the observed shortening.

		א נ	Preshot D	epth L,	Water D	epth (ft)		Shrinkage	Peak
Hole	Range (ft)	z	(ft); Sea	Floor	Preshot	PEACE	ΔD, ft	ΔD/ L . %:	Overpressure
	from GZ	ш	to C	to D	DI	D2	D2 - D1	Column to D	MPa
	6057.8	1	NA	NA	-	55.2	0	NA	1.0
05M-22	5538.6		NA	NA	,	76.0	0	NA	1.3
05R-21	5495.3	A	AN	NA	1	84.0	0	NA	1.3
07-20	1845.8		144.5	375	70.0	101.4	31.4	80	34
00T-19	1444.3	0	362.0	380	46.0	117.5	71.5	19	72
0TG-23	804.6	z	796.4	787	45.6	164.0	118.4	15	620
082-4	7.1	υ	1125.6	1068	13.1	198.7	185.6	17	069V
0CT-5	658.3		841.5	891	16.2	163.7	147.5	17	1690
0GT-9	1043.5	æ	1	1	16.0	134.8	118.8	I	228
0FT-8	1129.1	வ	593.8	611	15.4	130.8	115.4	15	172
0ET-7	1374.8	ស	477	775	18.4	106.9	88.5	11	85
0DT-6	1714.9	<u>G</u> .,	9.162	211	20.0	87.4	67.4	32	41
OAM-3	4510.2		NA	NA	I	108.0	0	NA	2.3
OAR-2A	4500.2		NA	NA	ı	110.5	0	NA	2.3
OAM-1 CAR-2	4458.4		NA	NA	ŧ	114.2	0	NA	2.4
0LT-14	2511.2	υ	-	1	132.5	139.7	7.2		13
OMT-15	2203.6	Q .	1	1	141.8	110.9	30.9	I	19
ONT-16	1827.3	0	93.0	207	130.9	135.1	4.2	2	34
0JT-12	1695.5	S	257.3	617	115.0	143.8	28.8	5	43
CHT-10	1462.2	S	446.5	626	124.6	137.3	12.7	2	20
011-110	1205.5		568.5	636	121.6	155.0	33.4	ç	136
0KT-13	988.7	84	730.0	664	101.7	164.7	63.0	6	269
0PZ-18	334.8	ш	1032.6	1043	46.3	201.9	155.6	15	>690
082-4	7.1	ស	1125.6	1068	13.1	198.7	185.6	17	>690
0UT-24	858.0	œ.	828.4	782	1.6	147.0	145.4	19	510

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TABLE 6-1. -- Column-height changes down boreholes at the OAK site. Contour C is the base of region of decrease in sonic wavespeed; contour D is the base of the region of downward displacement; GZ is ground zero.

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		5c						1114.6	e													(01114.0)	1114.6	
		Sb						1065.1													(1036.5)	(1063.0)	1065.1	(1025.0)
		5a	961.2		(1013.5)	(1020.1)	(1000.4)	1013.8	944.6	 	(925.0)	(925.0)	(925.0)	(918.2)	(1010.2)	(1013.5)	(83.8)	(0.166)	(987.3)	(0.086)	(974.5)	1000.001	B. CIOL	(925.0)
1		4b	765.2		(767.0)	766.5	(787.0)	847.7	7.997	 	(194.0)	(0.567)	(192.0)	(812.2)	(200.0)	6.107.)	(0.212)	(0.221)	(751.0)	(758.0)	765.8	6.608	847.7	784.0
in ft bs	ron	3b	552.4		552.0	548.3	610:0	701.2	572.2	 	(565.0)	(0.355)	(546.0)	(556.8)	(534.6)	(1529.7)	(6.7(5))	10.165	(1231.4)	(1562.0)	~ \$64.0 ~	723.5	701.2	(0,265)
E Depth,	HOL	3a	405.7	391.7	1.11	413.3	484.0	0.665	1.264		(419.8)	(410.0)	(0.795)	410.6	(8,585)	(6.676)	(1395.2)	-(E.Dec)	(• · E O •)	434.8	. e. jet	593.0	0.662	457.1
PEAC	PEACE	24	363.1	344.2	346.7	365.3	434.0		417.9		344.6	1320.4	(0.215.0)	355.6	(11.1)	(334.6)	(8.766)	(1350.01)	1(360.8)	0.375.0	411.6	568.9		407.0
		2c		290.1	262.7	274.7			368.4		272.0	294.7	E.162	310.0							1.926	ľ		1.0.676
		2b			216.2	6.652					[223.3]	220.6	219.0	,	(227.0)	(225.0)	238.6	238.0	213.3	274.4	232.9			
	Surface	•	ž	NA	445	426	(834)	1081	4706		(194)	(667)	231	٧N	(200)	(202)	(112)	(732)	(121)	(158)	766	(1089)	1081	(764)
	Hole	Base	1146.3	438.3	593.2	819.0	751.3	1803.9	1015.2	209.8	414.3	338.6	251.6	521.0	188.9	187.5	287.4	241.1	299.8	441.5	920.0	950.5	1803.9	498.1
	RANGE, It	to 082-4	6060.6	5498.2	1847.8	1446.1	805.5	0	654.3	1039.6	1125.0	1370.5	1710.5	4495.5	2517.3	2209.7	1833.4	1701.6	1456.3	1199.4	982.7	329.1	0	864.2
		Hole	00R-17	OSR-21	ORT-20	00T-19	0TG-23	082-4	0CT-5	0GT-9	0FT-8	OET-7	0DT-6	OAR-2A	0LT-14	OMT-15	ONT-16	0JT-12	OHT-10	017-11	OKT-13	0P2-18	0B2-4	0UT-24.
	Head-	ing	SW		α: 0	4 64	04	3 **	· თ	2			-	N N N	SE .	< (ء ر 	× C	- u	n u	0 o	4 60 	ده ۲	A MN

TABLE 6-2. -- Uniformity of horizons in OAK area. Parentheses () signifiesseismic data because borehole ends above depth shown; cross-hatchingcovers downward-displacement region; blank spaces signify missingvalues. Below sea level (given in ft) is abbreviated bsl.

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6-6

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the few tens of feet or less by which horizon depths vary (tbl. 6-2) have scant effect on the values of $\Delta D/\Delta z$ in Table 6-1.

SHORTENING OF CORAL COLUMNS BY DENSIFICATION: BOOKKEEPING

Let subscripts L and S refer, respectively, to the liquid and solid components of saturated coral. In a volume V of the mixture, let V_L and V_S be the volumes of the two components and ρ_L and ρ_S their densities. The mass, m, of the mixture is then equal to the mass of its liquid component $(=\rho_L V_L)$ plus the mass of its solid component $(=\rho_S V_S)$. Hence, if ρ denotes the density of the mixture, we can write:

$$\rho_{\rm L}V_{\rm L} + \rho_{\rm S}V_{\rm S} = m = \rho V$$

or

- .

where α_L and α_S denote the volume-fractions of liquid and solid in the mixture:

$$\alpha_L = V_L / V$$
; $\alpha_S = V_S / V$; $\alpha_L + \alpha_S = 1$ Eq. (2)

Using the last of Eqs. (2) to eliminate $\alpha_{\underline{L}}$ from Eq. (1), and rearranging, we get:

/olume Fraction of Solid in Mixture =
$$\alpha_S = (\rho - \rho_L) / (\rho_S - \rho_L)$$
 Eq. (3)

Hence, in volume V of the mixture, we find that:

Mass of Solid =
$$\rho_S V_S = \rho_S a_S V = V \rho_S (\rho - \rho_L) / (\rho_S - \rho_L)$$
 Eq. (4)

Now consider a vertical column of coral of unit cross-section. Let the column be divided into short vertical sections. A section of the column of height dh then subtends a volume V, and Eq. (4) -- with dh replacing V -- gives the mass of solid in that section. Summing over all sections of the column from a height z_0 to a greater height z, the total solid mass m_S between those heights is given by:

$$M_{\rm S} = \rho_{\rm S} - \frac{\rho - \rho_{\rm L}}{\rho_{\rm S} - \rho_{\rm L}} dh \qquad Eq. (5)$$

In Eq. (5), ρ_L , ρ_S , and ρ can all vary with height h in the column. Here however ρ_L (the density of sea water) is constant, while ρ_S can run only from about calcite's density to aragonite's (ρ_S can be set uniformly to the mean of the calcite/aragonite densities, with negligible error in m_S ; below). Thus,

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the measured density of the mixture, ρ , holds the key to simple subsidence in the OAK event.

With the key, goes a key assumption: The pre-shot density profile down any crater hole is the same as that found now in holes outside the crater ("control holes"), where the medium is almost unmarred. Then, taking z_0 [Eq. (5)] at a level in the column where coral has not been vertically displaced, the vertical thickness subtended by solid mass m_S [from Eq. (5)] in a control hole, is equal to the pre-shot thickness of mass m_S of solid in the crater hole. On that basis, the hypothesis of simple subsidence can be tested via its mandate to conserve the column's solid mass. For, the present thickness of that mass [also from Eq. (5)], subtracted from its pre-shot thickness, will give the actual change in height of its topmost particle -- <u>if</u> that change is due to simple subsidence.

In particular, if z [Eq. (5)] refers to the crater flocr, the change in question should equal the observed sea-floor lowering. Moreover, knowledge of the pre- and post-shot depths of horizons below the crater allows us a stronger result: By letting z_0 and z refer to any two horizons, Eq. (5) should give the same solid mass m_S pre-shot as now -- if the distance between horizons changed by means of simple subsidence. We therefore integrate upward from one and the same horizon R, both pre-shot $(z_0 = z_r^p)$ and now $(z_0 = z_r^n)$. When that is done (with a control-hole profile taken as "pre-shot"), a given solid mass m_S , reached at $z=z^p$ pre-shot, will be reached at $z=z^n$ now. For that solid (between z_r^p and z^p pre-shot), the difference $(z^n-z_r^n) - (z^p-z_r^p) \equiv \delta z$ specifies the change in thickness implied by the observed density-profile changes, if the solid moved only up or down. Thus, the meaning of measured density profiles for simple subsidence is shown by plotting δz (but with z increasing downward, not upward; i.e., with depth in place of altitude). Such plots tell how coral solid at any depth below the present OAK crater had its depth changed by the shot -- if simple subsidence caused the change.

DENSITY PROFILES, THEIR TREATMENT, AND DOWNWARD DISPLACEMENTS

Logs of back-scattered neutron and γ -ray intensity (see Melzer, 1986), and of gravity-field variations (Beyer, Ristvet, and Oberste-Lehn, 1986) furnished density profiles down boreholes in the OAK crater region ("neutron", " γ - γ ", and "BHG" profiles, respectively). On the reef-wise line (see second section), control-holes OOR-17 and OSR-21 were logged in all three ways, whereas neutron and γ - γ logs were taken in control-hole OAR-2A. There were no cross-reef control holes. On the crater's wing, however, the only reef-wise holes logged close to contour C or D were OQT-19 and ORT-20;¹ that was done by all three methods, save for neutron logging of ORT-20.

¹ That can be seen by comparing (1) the depths listed for contours C and D in Table 6-2 and (2) the density profiles presented in full in Appendix 6-1.

It was known before PEACE operations began at the PPG that we would have to look mainly to BHG for density profiles. Why? Because $\gamma-\gamma$ logs tell about the medium only within a few centimeters of our 4-inch-diameter boreholes, where the drilling disturbance is greatest; neutron logs "see" about 4 inches farther out (L.S. Melzer, conversations, summer 1987). By contrast, BHG logs give average densities out to about 10 times the vertical interval between readings (generally an interval of 25 ft for the boreholes in question); BHG densities are thus virtually free of man-made or natural local variations in the medium. It was not known, however, whether BHG logging could be done with useful precision under PPG conditions; doing so was a first, and a major PEACE Program success (see Beyer, Ristvet, and Oberste-Lehn, 1986, and Chapter 2 of this Report).

BHG aside, steel borehole casing that ran downward from the sea floor for 100 to 150 ft, interfered with $\gamma-\gamma$ logs; the tool was not calibrated for measurement in coral through such a pipe (L.S. Melzer, conversations, summer 1987). In view of that problem, and of changes to the medium from drilling, $\gamma-\gamma$ density profiles are probably reliable only at depths greater than a few hundred feet (where they match BHG profiles fairly well). Further, if neutron logs are to add density profiles to the BHG/ $\gamma-\gamma$ set, a way will be needed to calibrate the neutron tool for PPG coral (L.S. Melzer, conversations, early summer 1987). Thus, at present, density changes from the OAK event must be evaluated from BHG density profiles, augmented somewhat by $\gamma-\gamma$ profiles.

Copies of all BHG density profiles from OAK's reef-wise line are shown in Figure 6-2, and the $\gamma-\gamma$ profiles in Figure 6-3; all profiles appear at full scale in Appendix 6-1. For use in Eq. (5), each profile was fit by a piecewise linear function, an especially simple matter for the BHG stepprofiles: the linear coefficients are listed in Appendix 6-1, where the fits are also plotted. At full scale, the fits overlap the measured profiles everywhere, reproducing them about as closely as their finite line width allows. Those fits embody almost all the depth-dependence of the integrand of Eq. (5); the rest stems from the solid component's density $ho_{
m S}$, whose extremes lie within 4 percent of their mean (see preceding section). As measured, the variation of ρ_S over that small range is also no more than piecewise-linear with depth. Thus, at its worst, Eq. (5) calls only for integrating a ratio of two linear functions [since $\rho_S/(\rho_S-\rho_L)=1+\rho_L/(\rho_S-\rho_L)$] -- whence, down a given borehole, the solid mass mg is easily found in closed form vs. depth. When mg for a crater hole is equated to m_S for a borehole, however, the resulting equation for z^p in terms of z^n is transcendental. By taking ρ_S as constant over each of the many linear intervals of measured density ρ , we avoid that complication; m_S becomes (at worst) piecewise-quadratic in depth, and z^n-z^p becomes an explicit function of z^n . The results, plotted in Figures 6-4 through 6-7, are identical (when plotted) with those obtained by solving the transcendental equation; indeed, simply replacing ρ_S by its mean (2.821 g/cc), and ignoring its depth-dependence, makes no significant change in the figures. Full equations and details of calculation, including the fits to ρ_{s} , are presented in Appendix 6-2.

The dotted and dashed curves in our thickness-change figures (figs. 1-1 through 6-7) and in Appendix 6-1 speak to a subtler point in the treatment . density profiles: They make direct use of all horizon-depths measured for a given control-hole/crater-hole pair. Specifically, Eq. (5) was integrated from any one horizon to the next higher one in the given control hole. Starting from the same lower horizon in the crater hole, the zⁿ-value distant by equal solid mass $m_{\rm g}$ in the two holes, was computed from Eq. (5) (in the usual way; above) for each z^p -value. On reaching the next horizon in the control hole, z^n fell above or below -- but not on -- that horizon in the crater hole; the reasons: natural density-profile variations along the reatwise line, and sources of thickness-change other than simple subsidence. Integration proceeded nonetheless from that horizon in both holes, until the horizon above it was reached in the control hole -- and so on until density data gave out in one hole or the other. A full thickness-change curve was thus developed in sections, with the assurance that integration started for each section from the bottom of the same geologic (bin- or lithestratigraphic) layer -- and hence in as nearly equivalent material as possible in both holes. Dots track that curve in our plots of thickness change. The process was then repeated with the roles of crater hole and control hole reversed (i.P., going from one horizon to the next higher one in the crater hole; dashes limn that curve in our plots of thickness-change. The mean of the dotted and dashed curves -- a solid curve -- also appears in the plots. All three curves are clearly distinguishable on the right half of Figure 6-4 for example); note the borizontal step on the dotted curve, where integration to the upper surface of a layer in control-hole 008-17 took us past the corresponding surface in the crater hole (OPZ-18).

As the density profiles show (figs. 6-2, 6-3, and Appendix 6-1). log(in, began in each hole at a significant depth below the sea floor not at it. Calculated thickness changes (figs. 6-4 through 6-7 and Appendix 6-2) dust therefore be extrapolated up to the sea floor from the smallest depths the logs cover. With OSR-21 as the control hole, the gaps spanned by extrapolation at boreholes OQT-19 and ORT-20, respectively, come to about 5 and 6 percent of the present distance between the sea floor and surface B (downward displacement limit; see preceding section). Using control-hole of k-17, these figures grow to 23 and 21 percent -- large enough to have three people separately set reasonable upper and lower limits of extrapolation; arrows on Figures 6-6 and 6-7 mark the lines that gave the extremes of the six estimates and their mean.

The downward trend of every BHG-derived curve near its shallow of probably influenced all extrapolations (γ - γ curves go both ways; Appendix 6-2). If so, the bias can harlly be called a defect, given the trend's persistence. Indeed, it suggests forcibly that, down to 100 to 200 it below the sea floor, the medium is somewhat less dense now than pre-shot.

CONTRIBUTION OF SIMPLE SUBSIDENCE TO THE OAK CRATER

The solid curves of Figures 6-4 through 6-7 land Appendix 6-1 (proceeding) estimates of thickness change due to densitie tion below OAT crater. Extrapolating those curves to the sea floor producer the seam values listed as its in Table 6-3.



FIGURE 6-2. --Density profiles determined by borehole gravimetry (BHG) in the OAK crater area (Courtesy of L. Beyer, see Chapter 2). Depth below sea level (bs1) given in ft; bulk density in g/cm³. Inferred densities derived from gamma-gamma (Y-Y) logs shown as dotted lines. Asterisks (*) denote intervals where Y-Y logs are not available due to drillpipe.


FIGURE 6-3. -- Density profiles from gamma-gamma $(\gamma - \gamma)$ logging in OAK crater area (data from Melzer, 1986). Depth below sea floor given in tt.









FIGURE 6-6. -- Change in rock thickness from borehole-gravity (BHG) densities, assuming simple subsidence. Boreholes OQT-19 vs OSR-21 and OQT-19 vs OOR-17.



FIGURE 6-7. -- Change in rock thickness from borehole-gravity (BHG) densities, assuming simple subsidence. Boreholes ORT-20 vs. OSR-21 and ORT-20 vs OOR-17.

			High	Estimate	$\langle \mathbf{f} \rangle = \left\{ \begin{array}{c} .153, \text{ no bias} \\ .238, \text{ bias up} \end{array} \right.$		Best	Estimate	$\langle f \rangle = \begin{cases} .060, no blas \\ .146, blas up \end{cases}$	-	LOW	Estimate	$\langle f \rangle = \begin{cases}0.19, no blas \\ .077, blas up \end{cases}$	_
	ASE f: 0QT-19 0RT-20	α3	.17	.20	.19	.07	.24	.25	.20	EI.	.30	.30	.17	.20
Ξ δz/Δz)	to INCRE t bsl for at D for	f	.10	.23	.15	.47	.03	.20	05	.42	.02	.17	23	.36
: = 1 ft; f	BIASED D: 498 f Őz = 3 ft	δz	7.0	16.8	4.6	14.9	1.8	14.0	-1.7	13.1	1.2	11.9	-7.3	11.2
(Length Unit	D: 00T-19 0RT-20	۵3	.17	.20	.18	.07	.24	.25	.19	.14	.30	.29	.16	.21
	S MEASURE t bsl for t bsl for	f	.02	.16	.05	.38	05	.12	15	.32	06	60.	32	.26
	D: 423 f 442 f	δz	1.6	11.4	1.6	11.9	-3.6	8.6	-4.7	10.1	-4.2	6.5	-10.3	8.2
		Δz	71.5	71.5	31.4	31.4	71.5	71.5	31.4	31.4	71.5	71.5	31.4	31.4
		Pairs	19/17	19/21	20/17	20/21	19/17	19/21	20/17	20/21	19/17	19/21	20/17	20/21

TABLE 6-3. -- Column-height changes (in ft) due to densification. Borehole pairs shown in left-hand column; f denotes the fraction $\delta z/\Delta z$.

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If a column of material between D and the sea floor changed thickness via simple subsidence, then the thickness change δz computed from density profiles measured pre-shot (under our key assumption, see p. 6-8 above) and post-shot should equal the observed sea-floor lowering at the top of the column. The latter lowering is the column's actual thickness-change Δz (tbl. 6-3), whereas δz is the virtual change which densification provides, as computed from measured density profiles. For each control-hole/crater-hole pair, the ratio $\delta z/\Delta z$ appears in Table 6-3 as the fraction "f" of the sea-floor drop due to simple subsidence. On the wing of the OAK crater, the BHG profiles at hand (crater-holes OQT-19, ORT-20; control-holes OOR-17, OSR-21) tell a clear story: Only a small part of the sea-floor drop can be laid to simple subsidence. The highlights, subsumed in the f-values of Table 6-3, follow:

- (1). For the sea-floor drop, best estimates of the fraction (f) due to densification run from -.15 to .32, with a mean of .06.
- With each variable (some not yet discussed) pushed to a reasonable extreme so as to increase f, the minimum, maximum, and mean of f are .02, .38, and .15.
- (3). With each variable pushed to a reasonable extreme to decrease f, the least, greatest, and mean f-values are -.32, .26, and .01.
- (4). With further possible but unlikely increases in all δz-values (right side of tbl. 6-3), the best-estimate values of f [(1), above] increase by .10, .10, and .09, respectively; the f-values in (2) and (3), above, also increase by about those amounts.

Items (1) through (4) above cover systematic errors in determining the fractions by which densification changed column-heights. A major question remains, however, especially with so small a data-set: What confidence can be placed in these results?

CONFIDENCE ASSESSMENT

To fix levels of confidence, we look first at "thickness-changes" caused not by the OAK shot but by natural density-profile variations from one reefwise borehole to another. To that end, the profile from OSR-21 has been used as a crater profile with the one from control-hole OOR-17 (fig. 6-8) -- and vice versa (fig. 6-9). With OSR-21 as crater hole, extrapolation to the sea floor gives "thickness-change" extremes of -16.3 and -22.5 ft; with OOR-17 as the crater hole, the extremes become 19.8 and 30.4 ft. These are thicknesschanges that the two profiles would imply if simple subsidence, in a single coral column down to about 400 ft below sea level (bs1), turned one profile into the other. Thus, not only is density steadily higher in OOR-17 than OSR-21 to about 442 ft bs1, but, as forseen, ¹ the "changes" in question are a good deal larger than those due to the burst (the measured [δz ['s in tb1. 6-3

¹ Letters of July 12, 1984, and April 21, 1985, from author to Dean Oberste-Lehn, Research and Development Associates (RDA), and to Maj. Robert F. Couch. Defense Nuclear Agency (DNA), respectively.







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are ≤ 12.4 ft and average 3.0, 6.1, and .1 ft, respectively, for our best, high, and low extrapolations). With nature causing so much variation (presumably at random) in our calculated thickness-change δz , natural density variations dominate the δz -values. Indeed, for equal depths (442 ft) and the same control hole, absolute differences in δz between OQT-19 and ORT-20 come to <4 ft and average only 1.8 ft. Hence, treating all four of the resulting f-values as randomly distributed about a mean appears, at the least, a fair approximation (actually, three sets of four must be dealt with, owing to uncertainty in extrapolating δz to the sea floor). On that basis, the question of confidence turns to one of confidence in the estimated mean of a distributed random variable, and can be answered for a data-set of any size by standard statistical methods.

By reason of central tendency, and lacking data that would establish a precise distribution, we assume (as usual) that f-values are normally distributed. In each case (best, highest, and lowest extrapolation), the data of Table 6-3 then supply both a sample mean of f and a sample standard deviation. In computing the latter, however, it must be recognized that only three of the four f-values are independent. For example, given z in both OQT-19 and ORT-20, vs. z in OOR-17, we know z in OQT-19 vs. z in ORT-20; z in OQT-19 vs. z in OSR-21, then yields z in ORT-20 vs z in CSR-21. Since any three of the four f-values can be taken as independent, four estimates emerge of the standard deviation σ_3 ; two degrees of freedom enter the calculation of each (par for the standard deviation of three independent measurements). For each mode of extrapolation (high, best, low), every one of the four σ_3 -estimates is listed in Table 6-3 alongside the f-value omitted in its calculation; also listed are the drop Δz of the sea floor, and the part (δz) of the sea-floor drop due to densification.

For each estimate in Table 6-3 (high, best, and low) the mean <f> and a standard deviation imply a distribution of the true mean of the densificationfraction f. In its cumulative form, that "t-distribution" states the probability Prob(f>v) that the true mean of f (denoted f) lies above any stated value v. Since Prob(f>v) increases monotonically with σ_3 for a given sample mean $\langle f \rangle$, the largest of the four pertinent σ_3 -values was chosen to get a high estimate of Prob(f>v), and the smallest σ_3 -value for a low estimate; for the best estimate, the four σ_3 -values were averaged. With those respective values of the standard deviation σ_3 , we computed Prob(f>.5) and Prob(f>1) for each estimate_of $\langle f \rangle$ [high, low, and best; a value of about 1/2would be expected for Prob(f>1) if densification accounted for all of the seafloor drop]. The net result -- fundamental to the whole study_-- is that every case Prob(f>.5) ≤.05, with much smaller values for Prob(f>1). That is, the odds against densification having caused as much as half the observed seafloor drop are at least 20-to-1. The left side of Table 6-4 presents a detailed, though brief, summary of all key quantities. As for the right sides of Tables 6-3 and 6-4, the following applies.

The thickness-change curve for OQT-19 vs OOR-17 rises at its greatest rate at a depth of about 426 ft, from a minimum at 498 ft (fig. 6-6). The rise between 498 and 426 ft, and the curve's wiggles below 498 ft, may well stem from the medium's random reef-wise variations, rather than any error in estimating the depth of D at OQT-19. In addition, at ORT-20 (and elsewhere), non-zero downward displacement of a few feet might have been missed at D. Both hypothetical errors would act to decrease f; errors of opposite sign -about as likely -- would make our f-values too large. Nonetheless, to bias f

		D: 423 442	AS MEAS ft bsl 1 ft bsl 1	SURED: For OQT-19 For ORT-20				B Åz	IASED TO 498 ft b = 3 ft a	INCREASI sl for O t D for	с f 2т-19 окт-20	
			Ē ,	2		I			Ē>	5	144	1
Estimate	<f></f>	2 ³	ب	Prob	Lt.	Prob	<f></f>	03	ų	Prob	4	Proh
	.153	.07	8.31	.007	20.29	100.	.238	.07	6.52	110.	18.98	100.
High	.153	.15	3.89	.030	05.9	.005	.238	.16	2.85	.052	8.31	.007
	.153	.20	3.03	.047	7.40	600.	.238	.20	2.21	.079	6.44	č10.
	.060	.14	5.58	.015	11.92	.003	.146	.13	4.80	.020	11.58	.004
Best.	.060	.20	3.77	.032	8.06	800.	.146	.20	3.02	.047	7.28	600.
	.060	.25	3.06	.046	6.55	.011	.146	.25	2.43	.068	5.86	.014
	600	.16	5.50	.016	10.92	.004	.077	.17	4.30	.025	6.39	.006
Low	- 009	.24	3.63	.034	7.20	600.	.077	.24	3.03	.047	6.61	110.
	600	.30	16.2	.050	5.76	.014	.077	.30	2.44	.068	5.32	.017

TABLE 6-4. -- Likelihood that true mean f exceeds .5 or l, given a sample mean $\langle f \rangle$. Symbol f denotes the fraction of column-height change due to densification. For each estimate (high, best, low), the three standard deviations shown are are the greatest, mean, and least of the four values of Table 6-3.

as far toward higher values as common sense allows, we have assumed on the right of Tables 6-3 and 6-4 that: (a) the depth of D at OQT-19 is 498 ft, causing δz to increase by 5.4 ft (vs. OOR-17); (b) δz is also greater by 5.4 ft at D for OQT-19 vs OSR-21; (c) for ORT-20, the downward displacement is 3 ft (not zero) at 442 ft bsl, where the geologist's horizons place D; and (d) the 3-ft drop at D for ORT-20 is due entirely to densification at greater depths (contrary to all evidence above D). These assumptions lead to the values of f on the right of Figures 6-2 and 6-3, under the heading "BIASED TO INCREASE f". The main result (besides raising $\langle f \rangle$ -values by \sim .10): Prob(f >.5) \leq .08. Strengthened, therefore, is our central conclusion: Densification accounts for just a small part of the wing of OAK's crater.

LONG-TERM SETTLING

Within 67 days of the OAK shot (viz., B+67 days), the sea floor had been re-surveyed to ~3,000 ft from ground zero (GZ). Coupling that survey with PEACE bathymetry has brought to light notable changes in sea-floor depth between August 1958 and December 1984 (see Chapter 5 of this Report). Specifically, in that period, the sea floor sank by ~12 ft at ORT-20, 11 ft at OQT-19, and 4-1/2 ft outside the crater (3,000 ft southwest of GZ on the reefwise line). That cuts the respective sea-floor drops at ORT-20 and OQT-19 to ~20 and 60 ft seven to ten weeks after the event; i.e., "early". What those results mean for densification -- still our working hypothesis -- is perhaps plainest in terms of ORT-20.

Suppose first that the 1958 to 1984 sea-floor drop was caused by vertical settling of coral from the floor down to surface D; such densification is both simple and credible (any set settling below D, for example, would entail an error in present estimates of D's depth). The early sea-floor drop of ~20 ft at ORT-20 would take less densification to produce than the drop from pre-shot time to now (31.4 ft) -- but the column between D and the sea floor would have been 12 ft taller in 1958 than now, lowering its mean density. More precisely, the sea-floor drop δz due to densification would be less by 12 ft than in 1984 (see tbl. 6-3); the high, best, and low estimates of its value would all fall by that amount, making negative every f on Table 6-3 but one (for ORT-20). The opposite, less likely scenario has contour D move downward by the same amount as the sea floor above it. Above D, density profiles (and hence δz) are then unaffected; mean densification there stays in the small positive range shown in Tables 6-3 and 6-4 [$f=\delta z/(\Delta z - \Delta z_D)$; δz unaffected; equal changes in Δz and in the drop Δz_{D} at D] -- even if the drop Δz_{D} has other causes than densification.

Both extremes (D-depth unchanged vs. equal change at D and at sea floor) lead to the same fraction f of the sea-floor drop due to densification. For, if h_D is the increase in D-depth due to slow densification below D, then δz_D (the part of Δz_D due to densification) evolved to zero in 1984 from $-h_D$ at B+67 days (48 to 67 days after the OAK burst); similarly, if h denotes the increase in sea-floor depth due to slow densification above D, then (for coral above D) δz in 1984 becomes δz -h at B+67 days. On the densification hypothesis, we have $h+h_D = 11.8$ ft (the total sea-floor drop at ORT-20 from B+67 to 1984) for any value of h_D ; at OQT-19, $h+h_D = 11.2$ ft. Hence δz (densification's part of the total sea-floor drop by 1984) becomes δz -h-h_D at uay B+67 -- i.e., δz -11.8 ft for ORT-20 and δz -11.2 ft for OQT-19. Likewise, the sea-floor drop Δz (in 1984) becomes Δz -11.8 ft for ORT-20 at day B+67, and Δz -11.2 ft for OQT-19. The f's, reckoned as $\delta z/\Delta z$ for 1984, change to $(\delta z$ -h-h_D)/(Δz -h-h_D) for day B+67. Thus, with the 1958 to 1984 sea-floor drop h+h_D fixed (no matter how it is split into parts h and h_D due to densification above and below D), the change in f during that period is also fixed. Accordingly, at B+67, the f- and σ_3 -values of Table 6-3, and the probabilities of Table 6-4, becomes those of Table 6-5 below.

Table 6-5, like Tables 6-3 and 6-4, tells a clear and simple tale: As in 1984, the odds against densification having caused half or more of the seafloor drop measured at B+67, are \geq 18-to-1 in all cases; biased to favor densification, the odds remain \geq 16-to-1. At B+67, however, all values of $\langle f \rangle$ are negative, with a best estimate of -.30 (vs. .06 in 1984). Hence, had fvalues been as precise for B+67 as for 1984, Prob(f>.5) would have been a good deal smaller than .05; higher σ_3 's at B+67 blocked that (the largest were almost twice as big at B+67). Still, simple subsidence points to negative densification at B+67 (f-values $\langle 0 \rangle$, and it may actually have been negative then (dilatancy). More likely, though, the simple subsidence hypothesis is at fault; it is hard to believe that a medium with 40 to 60 percent porosity, even though fully saturated, would show sizable volume increases on loading and unloading either in uniaxial strain [peak overpressures: ~30 to 78 MegaPascals (MPa) (Table 6-1)] or thereafter.

Larger σ_3 's, and the increased scatter of f-values they reflect, take some explaining. They stem primarily from the reduced value, for ORT-20, of the column-height change that forms the denominator of f (a factor of 1.6 smaller at B+67 than 1984). Physically, small column-height changes can flag a breakdown of the f-criterion for measuring the part densification played in forming the crater. That measure makes sense only if the column-height change $(\Delta z-h-h_{D} \text{ in this case})$ is large compared to random ups and downs (standard deviation) in the part densification contributes to the change. Otherwise values of f for columns with small changes in height (changes adding little to the sea-floor drop and crater volume) will dominate the mean value $\langle f \rangle$ used to characterize the whole set of f's (including f's for holes with much larger changes in height). The problem can perhaps be finessed (below, last setion), but the true cure lies in computing f as a fraction of crater volume due to densification -- given axial symmetry, a sum (over all crater holes) of products $R\delta z$, divided by a sum of products $R\Delta z$ (R = horizontal range at a given crater hole). That fraction, after all, is the ultimate object sought. On present knowledge, its uncertainty would come mostly from its dependence on the choice of control hole for computing each dz. Here, however, we have density profiles from just two crater holes and two control holes. They yield too small a sample (2 ratios) to make such a criterion practical; the one adopted here, despite the drawback under discussion, permits more efficient use of those data (3 independent f's). Given profiles from a half-dozen or more holes of each kind, the volume criterion appears the better choice.

As for slow settling beyond the presently defined crater, pre-shot and 1984 contour maps show ~4 to 6 ft of it. So do the maps for B+67. The drop appears widespread as to direction, occurring (for example) at both 2-to-4 o'clock relative to GZ (north = 12 o'clock), and 6-to-8 o'clock. Here, however, it matters only if it means that the shot appreciably disturbed control-hole material. That is not at all likely for the following reasons:

		AS MEA	SURED		BIASED	TO INCREAS	Ef	
Pairs	Δ2	δz-(h+h _D)	f	03	δ2-(h+h _D)	Ţ	٤0	
19/17	60.3	1.6-11.2	16	.30	7.0-11.2	07	29	Hiah
19/21	60.3	11.4-11.2	00.	.27	16.8-11.2	60.	.26	Fist i mate
20/17	19.6	1.6-11.8	52	60.	4.6-11.8	37	.12	() 7 no biac
20/21	19.6	11.9-11.8	.01	.27	14.9-11.8	.16	.23	<f><f>= -05, bias up</f></f>
19/17	60.3	-3.6-11.2	25	.45	1.8-11.2	16	.43	Best
19/21	60.3	8.6-11.2	04	.40	14.0-11.2	.05	68.	Estimate
20/17	19.6	-4.7-11.8	84	.11	-1.7-11.8	69	.12	(-30, no biac
20/21	19.6	10.1-11.8	09	.42	13.1-11.8	.07	.38	<f><f>= {18, bias up</f></f>
19/17	60.3	-4.2-11.2	26	.58	1.2-11.2	17	.56	LOW
19/21	60.3	6.5-11.2	08	.53	11.9-11.2	.01	.51	Estimate
20/17	19.6	-10.3-11.8	-1.13	60.	-7.3-11.8	97	60.	41. No bias
20/21	19.6	8.2-11.8	18	.56	11.2-11.8	03	.53	<f>= -29, bias up</f>
					-	•	-	

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LIKELIHOOD THAT THE TRUE MEAN $\bar{\rm f}$ exceeds .5 of 1, given a sample mean <f>

			AS MEAS	SURED :				8	IASED TO	INCREAS	E f	
		- - 	Ē >.	2	, ju				~ <u>-</u>	.5	144	×1
Estimate	<f></f>	σ3	ب	Prob	۰	Prob	<f></f>	0 ع	4	Prob	ų	Prob
	17	60.	12.26	.003	21.44	100.	05	.12	8.07	.008	15.45	.002
High	17	.23	4.95	.019	8.66	.007	05	.23	4.20	.026	8.05	.008
	17	.30	3.81	160.	6.68	110.	05	. 29	3.30	.040	6.33	.012
	30	.11	13.08	.003	21.22	.001	18	.12	9.62	.005	16.66	.002
Best	30	.34	4.07	.028	6.60	110.	18	.33	3.58	.035	6.20	610.
	30	.45	3.10	.045	5.03	.019	18	.43	2.75	.055	4.76	.021
	41	60.	17.68	.002	27.38	100.	29	60.	14.76	.002	24.10	100.
Low	41	.44	3.60	.035	5.57	.015	29	.42	3.25	.042	5.30	.017
	41	.58	2.73	.056	4.23	.026	- 29	.56	2.45	.067	4.01	.029

TABLE 6-5. -- Column-height changes due to densification in OAK crater 48 to 67 days after burst. Borehole pairs given in left-hand column.

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- (1). Whereas 4-to-6 ft of subsidence appears to have occurred around most of the crater, the pre-shot map runs only about 1.2 crater radii; there, peak overpressure was greater by a factor of 5-to-10 than at controlholes OSR-21 and OOR-17.
- (2). As horizontal range increases, a steady decrease occurs in the sea-floor drop observed between 1958 (pre-shot) and 1984. On the reef-wise line of BHG logging, those drops run from ~120 ft one-fourth of the way to the crater's edge, to ~64 ft halfway there, to ~5 ft at the edge itself. They and their rapid decrease with range were doubtless both caused by the explosive loads the medium bore -- loads which also decreased rapidly with range. Like those loads, the sea-floor drop should be a good deal smaller at two crater radii than one -and the drop of 5 ft at the crater's edge is already small, whether it came about by densification or not.
- (3). Uncemented layers were breached in the crater's central region, opening new routes for leakage of water to the surface from great ranges; at a speed of only 1 cm/hr, water would have flowed ~7,500 ft by 1984. Driving such flow, however, would be gravity, just as it has tended over geologic ages to force water upward through the local fissures and passages present in coral. Balancing gravity over that time has been the ability of the solid skeleton to support vertical loads without transferring them to interstitial water; owing to those very loads, the strength of uncemented sand is >0. Gravity and strength act no differently now than in the past -- and the absence of detectable change in the separation of horizons in control holes argues that the balance between them remains where it was struck ages ago.

PIPING

During simple subsidence, skeletal coral replaces water that flows from it; since coral solids are denser than water, the medium then densifies, in accord with Eqs. (1)-(5). Yet, as discussed above, applying Eqs. (1)-(5) to the observed density profiles accounts for only some of the observed sea-floor drop; in material below the crater floor, density has increased by only a small fraction of the requisite amount. However, the finding that material hundreds of feet below the excavation crater had risen to the crater floor (see Wardlaw and Henry, 1986b; and Chapters 3 and 7 of this Report), suggests a way out -- namely, transport of solid particles by upwelling water. Any observed changes in density and column-height can be brought about by such "piping", given the right ratio of solid to liquid in piped slurry; for example, no density changes will be seen if the density of the slurry equals that of the pre-shot medium. In addition, of course, the right amount of material must be piped. On that point, the idea founders; evidence of substantial piping is limited to the central crater region.¹ There is also an implicit demand that piped solid be transported not just to the sea floor, but out of the apparent crater; that puts direct measurement of the amount of piped material beyond reach now. Nevertheless, piping was noted at OAK; some of its properties follow.

Eqs. (1)-(5) remain valid, but it is no longer useful to ask what preand post-shot heights are subtended in a column by a given solid mass m_s . Rather, with solid leaving the column, the mass of solid between two coral particles that remain in it will be different before the shot than after; moreover, the distance between them changes as both solids and liquid are lost. To compute the effects of both losses, let V denote a pre-shot control volume of the medium in which the following definitions apply:

 α = pre-shot volume-fraction of liquid in V

 β = pre-shot volume fraction of solid in V = 1- α

 $p_{\rm I}$ = density of liquid component

 ρ_{c} = density of solid component

 ρ = mean pre-shot density of mixture in V.

As on page 6-7 of this Chapter, it then follows that:

$$\alpha \rho_1 + \beta \rho_c = \rho$$
 Eq. (6)

To describe the post-shot state of the same material, let

 γ = piped-out fraction (volume or mass) of the liquid within V

 $k\gamma$ = piped-out fraction (volume or mass) of the solid within V

 $\bar{\rho}$ = present mean density of mixture not piped from V.

¹ For a cratering mechanism, a useful measure of significance lies in the fraction of the apparent crater's volume that can be laid to it. The piping observed at OAK crater occurred only within ~.4 apparent radii from GZ -- the central crater -- whereas the main PEACE problem is to account for the wing beyond the central crater. Piping will merit great attention if it can be shown, by tight quantitative arguments, to have produced something like half the wing's volume. By that standard, the fact that piping occurred can only suggest it as a possibly significant mechanism. The same holds for other observations as well, applying (for example) to any sand boils outside the apparent KOA crater; what their quantitative relation might be to the volume of KOA crater (let alone OAK's) is not at all obvious [mud boils also appeared above the Tatum salt dome after the SALMON event (Werth and Randolph, 1966, p. 3409) -- clear proof of piping, but piping played no role in forming SALMON's cavity].

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Since αV and βV are the respective pre-shot volumes of liquid and solid, the volumes of liquid and solid piped out of V equal $\gamma \alpha V$ and $k\gamma \beta V$. Hence, of the volume V, the piped-out fraction ϕ is given by:

$$\phi = \frac{\text{Volume of Mixture Piped}}{\text{Pre-shot Volume, V}} = (\gamma \alpha V + k \gamma \beta V) / V = \gamma(\alpha + k \beta) \qquad \text{Eq. (7)}$$

Likewise, the mean density of the remaining mixture becomes:

$$\vec{\rho} = [(1-\gamma)\alpha V \rho_{\gamma} + (1-k\gamma)\beta V \rho_{\gamma})]/[(1-\phi)V]$$

=
$$[(\alpha \rho_{L} + \beta \rho_{S}) - \gamma(\alpha \rho_{L} + k \beta \rho_{S})]/(1-\phi) - [\rho - \gamma(\alpha \rho_{L} + k \beta \rho_{S})]/(1-\phi)$$
 Eq. (8)

Using Eq. (7) to eliminate γ from Eq. (8), the direct result is:

$$\overline{\rho} = [\rho - (\alpha \rho_1 + k\beta \rho_S)\phi/(\alpha + k\beta)]/(1-\phi)$$

Slight rearrangement of this last equation makes it linear in $k\beta/\alpha$, whence

$$k\beta/\alpha = [(\bar{\rho} - \rho_{L})\phi - (\bar{\rho} - \rho)\phi + (\bar{\rho} - \rho)]$$

= $\frac{k\gamma\beta V}{\gamma\alpha V} = \frac{Volume of Solid Piped}{Volume of Liquid Piped} \equiv \frac{V_{SP}}{V_{LP}}$ Eq. (9)

The sludge density, $\bar{\rho}_{SL}$, follows from the ratio V_{SP}/V_{LP} :

$$\bar{\rho}_{SL} = (\rho_{L} V + \rho_{V} V) / (V_{LP} + V_{SP}) = [\rho_{L} + \rho_{S} (V_{SP} / V_{LP})] / [1 + (V_{SP} / V_{LP})]$$
 Eq. (10)

Values for the mean densities ρ and $\overline{\rho}$ come from density profiles measured, respectively, in control holes and crater holes. Also, for a vertical column of OAK coral, the fraction ϕ is just the change in columnheight, divided by the column's pre-shot height. Hence, from PEACE observations, Eq. (9) allows us to compute the ratio of solid and liquid volumes in piped material, and Eq. (10) its density, if piping caused the changes observed. Perforce, then, those quantities constrain the piping process, whereas γ and $k\gamma$ simply fix the unmeasurable total amount piped. For example, if no density changes occur ($\overline{\rho}=\rho$), then the volume and mass ratios implied by Eq. (9) will be those of pre-shot material, and piped sludge will have the same density as the rest of the medium -- in which case, gravity cannot cause it to be piped. However, if a shot raises the medium's density a bit (as the PEACE logs indicate for OAK), the resulting small pressure head can push sludge upward. To help quantify that push, estimates of the density of piped material have been made from PEACE measurements using Eqs. (7), (9), and (10).

In Table 6-6 below are recorded: (a) the mean densities (ρ and $\bar{\rho}$) measured for OOR-17, OSR-21, OQT-19, and ORT-20 from their shallowest common horizon, down to each of three others (the deepest at base D of the downward-displacement region); (b) the depths z and z_0 , respectively, of the top and bottom of the column to which each mean density refers; (c) the measured

column shrinkage, ϕ , between z and z_o for OQT-19 and ORT-20⁻¹ (together with the changes in depth Δz and Δz_o at z and z_o)²; (d) volume ratios implied by those data and Eq. (9), for the four control-hole/crater-hole pairs; (e) a mean density of piped material (also listed) follows from each volume ratio by Eq. (10), and with it (f) a "density decrement" $\rho_{\rm SL}$ - $\bar{\rho}$ (the difference between the densities of piped and remaining material). Evidently, subsidence by piping would require extruded material to have a bit lower density than that not piped. Note, however, that the residue's density $\bar{\rho}$ runs from slightly greater than that of the supposed sludge, to ~.45 g/cc less. That wide spread reflects sensitivity of the volume ratio [Eq. (9)] to random differences among borehole density profiles, when column shrinkages (ϕ) are <<1. Thus, the most consistent sludge densities and density decrements are obtained for the longest columns (third quartet of tbl. 6-6, running down to D at ~443 ft bsl for OQT-19 and ORT-20).

From the decrements in Table 6-6, it appears that slurry would be driven upward by pressures of about a tenth of the lithostatic head (mean decrement ~.2 g/cc), though the standard deviation of decrements is also that large (.21 and .26 g/cc; second and third quartets). At an upward acceleration of .1 g (decrement ~.2 g/cc), sludge would take ~11 sec to rise 200 ft in a wide, unobstructed pipe -- but there's more to piping than that.

OTHER CONSTRAINTS; HORIZONTAL PIPING

The densities in Table 6-6 apply to vertical columns 200 to 300 ft in height. Within such a column, single layers could have been driven by a density decrement as large as 1/3 g/cc. However, the path of sludge piped from the crater's wing leads first to the central crater, where lie nearly all the vents known to have guided solids from depth to the sea floor. That first path-leg has its pitfalls. For one, all horizons grow in depth along it; the horizons crossed by contour D at boreholes OQT-19 and ORT-20 (roughly 3a in fig. 6-1) run ~70 ft deeper at OTG-23, 800 ft from GZ (tbl. 6-2), and the sea floor lies ~55 ft deeper. Adding 55 ft of sea water and 15 ft (70 minus 55 ft) of coral makes the overburden ~11 percent greater than at the intersection of D with OQT-19 (or ORT-20). Along the horizon in question (H, say), the resulting overburden gradient opposes inward flow to the vents -- and while 11

¹ The "Volume of Mixture Piped", needed to calculate ϕ by Eq. (7), is equal to the change in depth Δz_0 of the column's bottom end, minus the change in depth Δz of its top end. The column's "Pre-shot Volume, V" is equal to the pre-shot depth $z_0 - \Delta z_0$ of its bottom end minus the pre-shot depth $z - \Delta z$ of its top end, where z_0 and z are the current (1984) depths of its bottom and top ends, respectively.

² Values of z, z_0 , Δz and Δz_0 were obtained for Table 6-6 from Table 7-4 of the present Report. For several horizons, the latter table lists both 1984 depths measured in crater holes, and estimates of pre-shot depth based on the full set of 1984 measurements (including horizon depths in control holes). Given a 1984 depth, the pre-shot depth of the same horizon was obtained for Table 6-6 by linear interpolation in Table 7-4 -- with the change in its depth equal to the difference between the 1984 and pre-shot values.

Pair	2 ⁰	8	Δzo	Δz	ф	ē	σ	V _{SP} /V _L p	- Psl	
19/17	309	220	35.0	38.9	.042	16.1	1.96	0×	٩<	×0
19/21	309	220	35.0	38.9	.042	16.1	1.89	.36	1.50	40
20/17	300	192	19.7	29.7	.085	1.91	1.96	4.66	٩<	0<
20/21	300	192	19.7	29.7	.085	1.91	1.89	.62	1.71	19
19/17	410	220	9.2	38.9	.135	1.94	1.94	1.04	1.94	00.
19/21	410	220	9.2	38.9	.135	1.94	1.89	.44	1.57	37
20/17	408	192	6.0	29.7	660"	1.94	1.94	1.14	1.98	.04
20/21	408	192	6.0	29.7	660*	1.94	1.89	.34	1.48	46
19/17	444	220	0	38.9	.148	1.95	1.94	.92	1.89	06
19/21	444	220	0	38.9	.148	1.95	1.90	.47	1.60	35
20/17	442	192	0	29.7	.106	1.95	1.94	.97	1.91	4 0
20/21	442	192	0	29.7	.106	1.95	1.90	.37	1.51	- 44

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TABLE 6-5. -- Piping hypothesis. Symbol \overline{p}_{SL} denotes density of piped material; densities in g/cm³; lengths in ft.

percent of overburden may be a small pressure, it is two-thirds or more of the total head available to pipe sludge from H. That head acts at the vents; slurry near them can, of course, be piped upward. At bigger ranges, however, their influence weakens relative to that of overburden. Hence, if we are dealing with a liquefied layer (all sludge), the denser layers above probably settle soonest near the vents, replacing piped material but pinching off the flow. Note also that slurry converges cylindrically as it moves inward, slowing its passage to the vents. The "aperture" available to it (proportional to horizontal radius) decreases by a factor of 2-1/2, for example, as slurry goes from 2,000 ft of radius to 800.

An unliquefied layer presents added bars to piping. For, in a layer with strength, unpiped material bears at least part of the overburden; the pressure that drives piping is smaller than the head that the density decrement would otherwise supply. Indeed, the slurry pressure may simply equal its own head, as in any drained unit; then no piping occurs. More generally, creep of the layer's strong component, like weakening induced by the blast, provides some impetus for piping -- but reduced from that which the full density decrement could furnish, and on a wholly aifferent time-scale. Indeed, creep can be so slow that almost no solid particles are entrained by piped water (simple subsidence), which may well have been the mechanism for settling between B+67 and 1984 (see preceding section). In addition, members with strength physically block piping; sludge has to flow between and around those solid parts. Such flow -- through a porous solid -- is described in simplest quantitative terms as diffusion, in accord with D'Arcy's Law, with flow rates set mainly by the medium's permeability. The lower the rates, however, the more solid settles out (under gravity) on its way to the crater's floor; further, at any given rate, entrained particles will not accelerate upward unless the drag on them exceeds their submerged weight.

These remarks suggest detailed calculations of upward/inward diffusion that have not been made, partly because the medium's post-shot permeability is poorly known, but more because PEACE disclosed no piping of note on the crater's wing.

Once at the surface, slurry particles would have to ride out of the crater on reef-wise currents of perhaps 1 knot (~1.5 ft/sec) (Halley and others, 1986, p. 5). During that half-hour trip, gravity would cause particles to settle; those with diameters >1/8 mm would drop an estimated 100 ft or more along the way (Stokes flow), and hence would leave the crater, if at all, only by other, slower means. The same forces of drag, weight and buoyancy also act on the particles during their rise to the crater's floor; treating them again as isolated spheres, the buoyancy and drag of water rising 200 ft in one hour (1/18 ft/sec) can move them only if their diameters are <1/8 mm, while for 10 and 100 hours of rise, respectively, the critical diameters are 3/80 and 1/80 mm. These estimates are rough, since the particles are not spheres (nor do spheres bound drag \ddagger mass), they and their wakes overlap [increasing drag \ddagger mass if they do not clump (Soo, 1967, Ch. 5)], and they rise in ragged, twisty channels (not straight, free streams) that may be <10 diameters wide in places.

Fissures, and cones of debris containing coral fragments raised hundreds of feet, were seen in the central region, where coral was most damaged (Halley and others, 1986; Slater, and others, 1986). Piping accounts neatly for that, and much of it could have occurred in a few seconds or less. For, with

overpressure at the .1-MPa level, burst-induced sub-crater pressures up to ~100 MPa would furnish the required vertical stress gradients. Relief of those pressures would be rapid, requiring a volume increase of <1/2 percent for decay to .1 MPa (from 100 MPa). Though slight in relation to crater volume, enough material would be extruded in such an expansion to make impressive deposits on the sea floor, and cloud the reef currents that cross the crater.

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DENSIFICATION: SUMMARY AND CRITIQUE

Borehole cores have let geologists fix the depths of many layer interfaces ("horizons") below and outside the OAK crater. Further, in two holes on the crater's wing, gravimetry has furnished density profiles down to horizons not moved by the OAK burst; to the same horizons, but well outside the crater, borehole gravimetry (BHG) also has given two density profiles. All four holes lie on a curve roughly parallel to the reef. There, in pristine coral, geology argues for random density-protile variations about some mean. The present coral medium formed in about the same way at different points on any one of a set of curves roughly parallel to the reef. Moreover, PEACE cores and density profiles support the idea of such variations about a mean. Treating the far-field pair as pristine profiles then yields density changes due to the burst, from depth to 20-100 ft below the crater floor. From those changes come the downward displacements that densification implies for sub-crater coral, vs. depth, and for the crater floor itself. Comparing the latter displacements to actual sea-floor drops yields the result that densification played but a small role in forming the crater's wing.

Except for converting density profiles to downward displacements, significant uncertainty attends each step noted:

- 1. The general increase of coral density from lagoon to ocean is a source of systematic error in the measured profiles. Specifically, prior to OAK, departures from the mean density profile would have been random along the curve on which boreholes were supposed to lie. That curve is not known precisely. Actual boreholes therefore depart from it, but are about as likely to fall on one side of it as the other. Hence, given the oceanward density gradient, the general effect of such misplacement is to increase the differences among measured profiles. The unlikely opposite result, however, is more apt to have occurred in our four-profile set than in a large set. The scatter of profiles would then have been underestimated.
- 2. A horizon's drop (or rise) by a few feet could have escaped notice. That holds for the "unmoved" horizons above which we reckoned density-change effects in the two crater-wing holes.
- 3. From the shallowest point of BHG logging in a given borehole to the sea floor, horizon-depth changes due to densification were estimated by extrapolation from below.
- 4. Limits of precision render BHG-measured densities uncertain, but by < ± .02 g/cc. Further, BHG densities are averages over such large regions that the effects of them of local site inhomogeneities (vugs, etc.) are believed negligible (for PEACE, a great advantage of BHG over other methods).</p>

- 5. The sea-floor drop of implied by density changes down a borehole, divided by the actual sea-floor drop at the hole (Δz), measures the contribution of densification to the crater. That ratio (f), however, gives too much weight to holes where the actual drop is small. Further, though f is a random variable, its distribution may not be near-normal (as assumed).
- 6. The largest values of δz (vs. depth) come from pairing the two wells outside the crater, just 560 ft apart ("control-holes," ~5,500 and 6,000 ft from GZ). Moreover, paired with the same control hole, the two holes 400 ft apart in the crater's wing (~4,000 ft from the control holes) have much the same curve of δz -vs.-depth. Nature, not the OAK burst, thus appears the chief source of variation among the four density profiles; our signals (density changes due to OAK) were buried in noise (rendom natural differences in density). As a result, the likelihood that the crater wing formed by simple subsidence could be assessed using three independent values of δz (the maximum from four profiles) -- despite our having just two profiles from the wing. Strictly, that can be correct only as f, divided by its standard deviation, approaches zero. The crater-wing profiles show some densification (f>0), however, and may differ systematically therein owing to their different ranges.

Caveats 1 through 6 forced us to assess confidence in the overall finding of low densification (δz (.1 Δz). To that end, our data base and calculations were altered (within reason) to maximize f:

i) Since oceanward density gradients could have acted to reduce differences among our f-values [caveat 1 above], the standard deviation of f was assigned the largest value found from the data (of the four deviations at hand when three of four f's are independent).

; i) For each borehole, δz was set at the sea floor to the highest δz -value found by extrapolating δz -vs.-depth to the floor [point 2 above].

iii) Unseen displacement of the shallowest "unmoved horizon" D in a given borehole would probably have been downward. Each δz -value from item (ii) above was increased by 3 ft or more to offset such an error [caveat 3 above].

iv) Adding offsets (iii) directly to öz-values credits the unseen drop of D entirely to densification below D -- even though densification accounted for just a small part of the horizon-drops observed above D (in the crater's wing).

v) On the PEACE data, the tendency of f to place undue weight on holes with small sea-floor drops [point 5 above] led to high -- but accepted -- values of both f and its standard deviation (ORT-20 has smaller Δz -values than OQT-19 and higher f-values).

The overstatement of f flagged by item (v) looks correctible (next section). That correction would probably be cancelled, and then some, if no appeal were made to the limit where f, divided by its standard deviation, tends to zero (item 6 above). How to avoid that limit without giving f another strong ad hoc lift is not clear. As it is, each upward bias lent to f appears within reason. Having them all act at once to make high f-values likely does not. But even so, only minor densification results.

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CONCLUSIONS

If the wing of the OAK crater resulted from densification, then the seafloor drop at a wing-station, W, can be computed exactly from density profiles before and after the shot in a vertical column below W. No such profiles were measured pre-shot. That heightens the problem of reading density increases due to OAK through the noise of random natural variations in density. Below the crater's wing, those variations were found to dominate shot-driven changes in density. More important, however, the noise level proved low enough to admit a clear answer to the main question posed -- on the wing of OAK's crater, most of the sea-floor drop had causes other than densification.

As a best estimate, 6 percent of the sea-floor drop on the crater's wing can be laid to density increases caused by the burst. That figure follows from profiles down two crater holes and down two control holes outside the crater -- profiles that yield four estimates (3 independent) of the fraction, f, contributed by densification to the sea-floor drop. Each of the four has a high, best, and low value, depending on how a gap in data just below the sea floor is bridged (fig. 6-10). To be sure, the sample is small, but its size has been taken into account in assessing confidence in the mean of f. The results: The probability that densification caused half or more of the seafloor drop is <.1. That result holds even if the main parameters of the calculation are all varied at once (each within reasdonable limits) so as to increase f. The PEACE density profiles could be of course atypical, but, at most, that observation only supports measuring more profiles; with the data at hand, the results are as stated.

Extant maps show that, in the crater wing, the sea floor sank appreciably between August 1958 (a few months after OAK) and December 1984 (PEACE). The crater was therefore significantly shallower in 1958 than now. By the same token, given simple subsidence, the medium was notably less dense (on average), from the base of the region of downward displacement to the crater floor. Combining that slow sea-floor drop with the PEACE density profiles leads to a best estimate of $\sim -.2$ for f at August 1958 -- and again (by chance) the probability that densification caused half or more of the seafloor drop at that date is <.1.

The same data base of sea-floor maps and PEACE density profiles also yields mean values for the density of materials piped up to the sea floor, if the piping hypothesis is correct. From those values, the densities of piped and residual material differ by an average of $\sim .2$ g/cc, but with a standard deviation at least that large. A density difference of .2 g/cc can drive piping, but weakly -- and the chain of events leading to transport of piped material out of the crater has many weak links (e.g., it appears that particles >.1 mm in diameter will settle before they can exit).

The statistical grounds for assessing densification probably can be strengthened, using only extant data. Given the cost of the data, that should be done. Specifically, both the sea-floor drop (Δz) and the part of it due to density changes (δz) can be expressed as fractions of the pre-shot depth to D. The probability that the latter fraction exceeds half (say) of the former can then be computed, using standard deviations supplied by PEACE data. A BHG profile from the central region could also be added to the present set, but not without giving a further strong upward bias to the estimated extent of



FIGURE 6-10. -- High, low, and best estimates of the fraction contributed to sea-floor drop by densification at boreholes 0QT-19 and 0RT-20.

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densification. In addition, the plausibility of the piping hypothesis should be examined further. With densification as an unlikely mechanism for forming OAK's crater-wing, however, plastic flow appears to offer the simplest and most plausible explanation for it. Whether that explanation will withstand close scrutiny is unclear; flow is well understood in principle, but not much is known about the displacement field around a flow crater.

ACKNOWLEDGEMENTS

For the PEACE Program data herein, and much kind assistance besides, my sincere thanks go to Larry Beyer, Scott Blouin, Bob Henny, Woody Henry, Steve Melzer, Dean Oberste-Lehn, John Peterson, Byron Ristvet, Ed Tremba, and Bruce Wardlaw. Larry, John, Woody, and Byron, and (above all) Steve and Bruce are due special thanks for extended discussions of both the data and its implications; so are my co-workers Neil Perl, Jim Workman, and Ken Burrell, for reducing PEACE data to forms from which conclusions more readily follow.

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APPENDIX 6-1

This appendix contains (1) all PEACE density profiles measured in the OAK area, with (2) plots of the fits made to those profiles as part of the work reported here. Also presented are (3) tables of coefficients for the piecewise linear function used to fit all profiles. That function is defined as follows:

$$\rho = [(z-z_{j}) \rho_{j+1}^{-} + (z_{j+1}-z) \rho_{j}^{+}] / (z_{j+1} - z_{j}); j = 1, 2, ..., J \qquad Eq. (11)$$

where ρ and z denote density and depth, respectively. For the BHG profiles [received as step-functions from L.A. Beyer, written communication, May 15, 1987); see Chapter 2, this Report], ρ_j^+ and ρ_{j+1}^- have the same value ρ_{j+2}^+ . Otherwise, $\rho_j^- = \rho_j^+ = \rho_j$, and (p_j, z_j) gives the coordinates of an endpoint of either two or one straight-line segments of the complete function. Specifically, for $j \neq 1$ or J, a segment runs from (ρ_{j-1}, z_{j-1}) to (ρ_j, z_j) , and another from (ρ_j, z_j) to (ρ_{j+1}, z_{j+1}) ; the single segment for j=1 runs from (ρ_1, z_1) to (ρ_2, z_2) , and the single segment for j=J connects (ρ_J, z_J) to (ρ_{J+1}, z_{J+1}) .

The measured BHG profiles, in graphic form, comprise the first exhibit below (figs. 6-11 to 6-16). In each case, for ready comparison, a graph of the density-function fit to a given profile [Eq. (11)] is shown next to it, with the pair on identical scales. Then, in exactly the same format, a set of figures (6-17 to 6-26) follows in which appear all the profiles derived from γ - γ logging, together with the density function fit to each. Next, on a single page (tbl. 6-7), come all the (ρ_j, z_j) -points that specify the functions fit to BHG profiles (points supplied by the tables of Chapter 2 of this report). A corresponding table for fits to all the γ - γ profiles comes last (tbl. 6-8). The latter table was compiled by measuring coordinates from the profiles themselves, having overlain them with thin graph paper; thus, at the outset, our measures of density and depth, denoted "DIV" in the tables, were a pair of coordinates read off graph paper. Conversion was made from DIV to g/cc, and from DIV to ft, by means of the following formulas:

density $(g/cc) \approx Q + (DIV-Y_0)/S$; depth $(ft) \approx A + B(DIV-X_0)/C$ Eq. (12) Values of Q, Y₀, S, A, B, X₀, and C are given for each profile in Table 6-8.



BORDHOLE CRAVITY SURVEY: HOLE OOR-17



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FIGURE 6-12. -- Left: profile of density vs. depth from BHG logging in control hole OSR-21, as received. Right: plot of broken-straightline fit [Eq. (11)] to profile at left. The left- and right-hand plot scales are identical.

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BOREHOLE GRAVITY SURVEY: HOLE OPT-20

FIGURE 6-13. -- Left: profile of density vs. depth from BHG logging in crater hole ORT-20, as received. Right: plot of broken-straight-line fit [Eq. (11)] to profile at left. The left- and right-hand plot scales are identical.



BOREHOLE GRAVITY SURVEY: HOLE OQT-19

FIGURE 6-14. -- Left: profile of density vs. depth from BHG logging in crater hole OQT-19, as received. Right: plot of broken-straight-line fit [Eq. (11)] to profile at left. The left- and right-hand plot scales are identical.

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BOREHOLE GRAVITY SURVEY: HOLE OTG-23



FIGURE 6-15. -- Left: profile of density vs. depth from BHG logging in crater hole OTG-23, as received. Right: plot of broken-straight-line fit [Eq. (11)] to profile at left. The left- and right-hand plot scales are identical.

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FIGURE 6-16. -- Left: profile of density vs. depth from BHG logging in crater hole OPZ-18, as received. Right: plot of broken-straight-line fit [Eq. (11)] to profile at left. The left- and right-hand plot scales are identical.

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BOREHOLE COP-17: GAMMA-GAMMA LOGGING

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FIGURE 6-17. -- Left: profile of density vs. depth from γ-γ logging in control hole OOR-17, at .70 times the scale of the plot as received. Right: plot of broken-straight-line fit [Eq. (11)] to profile at left. The left- and right-hand plot scales are identical.

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BOREHOLE ORT-20: GAMMA-GAMMA LOGGING



FIGURE 6-19. -- Left: profile of density vs. depth from γ-γ logging in crater hole ORT-20 at 1.0 times the scale of the plot as received. Right: plot of broken-straight-line fit [Eq. (11)] to profile at left. The left- and right-hand plot scales are identical.



FIGURE 6-20. -- Left: profile of density vs. depth from Y-Y logging in crater hole OQT-19, at 1.0 times the scale of the plot as received. Right: plot of broken-straight-line fit [Eq. (11)] to profile at left. The left- and right-hand plot scales are identical.



The left- and right-hand plot scales are identical.

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BOREBOLE OB2-4: CAMMA-GAMMA LUGGING



FIGURE 6-22. -- Left: profile of density vs. depth from Y-Y logging in crater hole OCT-5, at 1.0 times the scale of the plot as received. Right: plot of broken-straight-line fit [Eq. (11)] to profile at left. The left- and right-hand plot scales are identical.

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FIGURE 6-23. -- Left: profile of density vs. depth from γ-γ logging in control hole OAR-2A, at 1.0 times the scale of the plot as received. Right: plot of broken-straight-line fit [Eq. (11)] to profile at left. The left- and right-hand plot scales are identical.

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FIGURE 6-24. -- Left: profile of density vs. depth from γ-γ logging in crater hole OIT-11, at 1.0 times the scale of the plot as received. Right: plot of broken-straight-line fit [Eq. (11)] to profile at left. The left- and right-hand plot scales are identical.







FIGURE 6-25. -- Left profile of density vs. depth from Y-Y logging in crater hole OKT-13, at 1.0 times the scale of the plot as received. Right: plot of broken-straight-line fit [Eq. (11)] to profile at left. The left- and right-hand plot scales are identical.



BOREHOLE OP2-18: GAMMA-GAMMA LOGGING

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FIGURE 6-26. -- Left: profile of density vs. depth from Y-Y logging in crater hole OPZ-18, at 1.0 times the scale of the plot as received. Right: plot of broken-straight-line fit [Eq. (11)] to profile at left. The left- and right-hand plot scales are identical.

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 TABLE 6-7. -- Endpoints of segments of piecewise linear fits [Eq. (11)] to

 density profiles from borehole gravity surveys.

	Pf	ACE BOREHOLL	704-17			P	CALL BOALHOLL	004-17	
J	0161714E0 019184 # 18191	DIWITIZLC DE44111, T 10141	(D40U7En 02974 1771	COMPUTEE Newgitt Ibm/CCI	J	DIGITIZE DEPTH, 4 (DIV)	014333721 C145175-3 10143	CUMBLITE DEPTH IFT1	COMPUTE: DENSIT: COMPUTE:
	5.00	23.30	72.72	1.903		74.00	42.00	352.17	4.001
÷.	5.00	43.50	74.44	1.907	1-5	70,90	24,10	356.84	1-917
2	5,70	22.5C	74.31	1.896	142	71.59	24,50	358.19	1,920
5	6.74	82.40		1.000	1*5	74.29	42,50	\$41.43	1.447
	1.54	12.50	42.41	1.447	240	72.64	25,98	343.33	1,931
	7.20	46.00	90,78	1,950	1.00	73.54	47.05	367.22	1.967
,	10.04	44.80	44.22	1.929		73.84	24.00	368.51	1.915
1.	19.28	23,20	73.44	1.97	110	74,48	26.58	371.00	1.950
	12.00	24.30	102,01	2.931	142	75,20	25,68	374.53	1.935
13	12.60	49.20	106.75	1.919	225	*5.50	40,30	375.01	4.851
15	1.04	44.20	131.41	1. 414	115	76.59	20.00	SA0.11	3.846
14	34.70	25.00	114.47	1.946	116	76.70	25,10	380.97	1.934
17	13.00	4	117.15	1.76*	117	78.80	24,20	384.43	1.919
	15.65	4	110.74	1.927	119	78.50	25,30	348,11	1.935
	16.01	24,45	120,01	1.422	140	80.00	29,50	19. 80	2.01
	14.90	41.70	123.88	1.475	142	81.00	26.00	399.46	1.950
43	17.30	56.00	125.60	1,950	143	61.40	45.03	401.10	1.935
	10,20	23,30	130.74	1.000	149	82.59	25,80 24.00	404.13	1,940
- 46	19.70	16.00	133.92	1.777	146	83.80	23,77	.04.86	1.910
27	21.20	18,60	192,37	1.825	147	83.70	25,80	411.0*	1.946
29	24.00	22,20	154.40	1.884	147	84.70	22,00	13.37	1.501
	24.10	23.00	154.83	1,933	1.50	85.00	16.00	416.66	1.791
21 22	25.00	29.00	160.47	1.915	121	86.00	19.0P	420,46	1.851
	25.60	25.20	161.20	3,936	2.33	#7.3U	**.50		1.924
	26.30	25.30	169.27	2.021	144	87,94	24,70	*2* 1*	1.921
36	37.94	29.00	171.17	2.002	136	80.60	29.50	432.13	2.010
37	28.30	34,80	172.09	2.102	137	89.00	27.80	433.05	1.961
30	27.69	38.08	178.48	2.017	138	87.30	21,20	435,14	1.667
	30.40	38.70	101.92	2.031	148		26,50	**1.59	1,950
1	50,80	29.50	183.64	2.010	191	91.20	25.20	**3.31	1.936
	31.89	25.30	187,94	1.938	148	92.30	48.46		1.991
	34.14	47.20	199,23	1.971	1**	92.84	29,60	150.19	4.012
46	54.20	27.30	198.26	1.976	195	45.60	26,30	*53,63	1,435
47	34.00	24.00	206,00	1,915	197	94.78	46.00	158.36	1.950
	37.20	29.40	211.15	1.929	148	116.40	26.00	551.65	1.950
20	59.00	49.80	218.05	2.016	120	159.00	32.00	540.05	2.054
21	39.80	31.20	222. 11	2.640	121	123.50	55.00	542.18	2,071
- 33	41.37	4 . 20	224,76	1.471	125	124.00	31,80	587 84	2.010
24	41.70	25,89	231.35	1.946	124	126,00	\$3.40	592.93	2.078
27	43.50	28.00	234.24	1.98-	133	126,70	51.60 17 80	395,94	2.047
37	- 5 . 6 0	26,80	230.67	1.***	197	120.04	44,80	601.52	8.102
38		48.00	243.40	1.984	124	129.00	35.BC	603.82	2.111
	**,20	27,30	245,55	1.972	159	150.50	\$1.00	411,41	2.036
	.7.89	24.34	255,29	1,993	101	131.80	27.00	617.86	1.967
• 3	47,10	27.60	253.72	1.978	143	134.80	48,90	622,16	2.000
	67.20	47.76	258.45	1.9.9	194	133.00	31.00	623.02	2.036
	.8.70	\$7,00	260.60	1.967	1.00	134.10	49.30	627.75	2.087
60	50.00	29,20	264.89	2.005	167	134,60	42.00 31.4	629,90	2.054
. 9	50.64	50.40	765.76	1.026	147	133.50	31.60	633.77	2.047
/1	51.00	26.00 47.00	270,64 279,42	1.967	1/0	135.70	32,00	634,63	2.034
14	82,00	23.50	. 74 , 78	1.907	1/2	137.64	27,48	4+0.22	1.474
- /3	54.84	42.30	277.79 279.88	1.852	175	157,50	29,30	642.37	2.807
15	55.90	20,80	282.95	1.846	175	130.70	30,00	647.53	8.017
	54.10	24.20	283.81	1.917	11	139.30	33.70	650 11	4.065
1.	55,59	25.30	289.83	3,930	176	140.80	32.80	676.55	2.067
	56.20	24,28	292.04	1.919	1.14	191.20	34,30	\$38.27	2.093
	56.90	44.00	295.45	2.088	100	141.60	\$2,00	660,85 664 34	2.054
	37.30	36.00	297.57	2.125	142	1	\$5,00	470.31	4.106
0) 84	38.00	22,40	500,58 \$01.44	2.001	1.03	145.20	\$0.50	675.67	2.020
	57.20	\$4.30	\$05.74	4.093	1*5	146.30	31.80	640.20	2.050
	60,00 60.40	25.00	309,1A 571 74	1,933	106	147.00	\$4,70	683.21	2.100
•	41.04	46.00	513.48	1,950	107	148.20	29,0C 40.30	688.37 691 14	2.002
	61.50	26.40	315.65	1.957	199	147.80	30.30	695,25	2.024
90	61.60	27,70	516.06 318 8*	1.979	190	156.00	\$1.50	696.11	2.045
	42.50	28,80	319.92	1	171	150.00	69.30	679,55 701,70	2.034
*5	63.40	50	323,79	2.010	193	152,70	<8.20	707.72	1.980
	63.30	47.50	528.95 531.44	1,975	144	154.20	51,80	714.16	2.050
	66.00	26,50	334.97	1.950	196	155,10	34,50	718.03	2.097
97	66.9U 67.8U	26,80	338,84	2.964	197	157.04	29.50	726,20	2.007
	44.30	28.60	344,86	1,995	199	158.30	29,30	731.79	2.079
140	69,50	26.00	350,02	1,950	240	1.0.40	\$3,50	760.82	2.060

TABLE 6-8. -- Endpoints of segments of piecewise linear fits [Eq. (11)] to density profiles from gamma-gamma logging. Table continues on succeeding pages. Data given for boreholes OOR-17, OSR-21, ORT-20, OQT-19, ORZ-4, OCT-5, OAR-2A, OIT-11, OKT-13, and OPZ-18 in OAK crater and KAR-1 in KOA crater.

6-57

and every size and the second

		TEACE BORTHOLD	008-17	
•	D161114EC D161114EC 10141	10141 Df #8144+ 4 DJ #1115FD	COMPUTER DEPTI IFTI	CUMPUTE: "LWGŽTT (VM/LE)
403	164.04	47.30	747.79	1.972
***	162.40	86.90	751.14	4.826
204	154,00	35.85	746.30	8.117
	165.34		741.04	2.135
408	166.30	36.08	764,84	2.123
204	166.70 167.80	85.7D	767.90	4.11*
211	150.00	30.40	.73.44	2.024
413	169.50	31.30 31.30	775.68	2.042
214 617	167.80 170.60	30.00	201.25	2.01.0
416	171.00		746.14	4,000
418	174.60	32.00	188,54 793,27	4.106 4.038
684 686	179.60	36.70	747 57	2.135
	1	26.30	M02 30	4.150
443	175.84		834.8A 807.83	2,022
245	1	20.00 22.76	810.47	4.002
646	179.00	33.80	822.51	2.085
648	101.00	34.30	928.32	2.067
430	181,80	46.80 33.81	832.82	2.123
231	103.10	\$2.20	838.41	2.037
	185.60	41.70	844.43	2.130
633	144.44	45.80	838.40	4.471
436	187.5# 148.80	43.50		8-114
130	189.50	10.00	865.95	8.019
	190.80	41.00	847.22	2.07 <u>1</u> 2.834
	191,96	48.00 40.30	876.84	2.157
243	194,20	42.88	844.15	2.047
244	196.60	\$1,70	890.00	2.227 2.048
e* 7	199.20	42,38	\$02.47 \$07.63	2.099
299	199,70 281,30	42.30	909.74	2.039
250 ⊀⊇1	202.30	30.90	920.96	3.024
	203.90	30.8p	***.11 *24.12	4.016
	201.00	22.30	932.36	8.048
676	204.30	35.20	\$38,15	2.892
	201_81 408_80	14.10		2.090
234	206.64	37.00	947.18 948.84	2.118
4+1	207.70	37.90 37.90	958.42	2.136
443	210,30	36,00	953.63	8.123
485	233,80 211,70	**.26	758.36	2.234
446	212.90	49.50	961.37 964.81	2.235
	214,50	27.80	967.82	1.981
463	513.00 513.00	34.48	979.00	2.095
471	217.70	20.10	987.36	5.051
413	219.00	22.00	989.3 <u>1</u> 992.74	2,071
215	520.50	<7.50 28.80	994.90	1.976
415	220.70	30.06	1000.06	2.157
478	221.80	35.00	3803.07 3824.79	2.140
	283.00	37.50	1006.31	2.149
245	223,40	23.50	1011.67	4.991
283 284	225.50	23.00	1019.44	1.896
205	226,60	24.00 24.70	1020.27	1.984
285 287	226.8V 226.8V	76.U0 62.ND	1025.43	1.950
674 694	227.50	20.05	1028.54	5,005
270	231.00	50.80	1041.76	2.044
472	231.50 232.04	28.60 27.80	1045,63	1. **5
693 674	232.50	48.00	1050,74	1.767 1.984
295	235.70	20.20	1052.94	2.014 1.988
297	235.00	48.10 49.30	1058.53	1.946
278 279	236.20 237.60	27.80	1066,70	1.981
340	239.20	34.00	1073.50	1.967 2.088

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		PEALE BO	DREHULL	104-17	
J	576111451	Diel	1740	ONDITIO	- Denistry
	64 8) - 1	01 h 51	11.1.1	JEPSH	GINELTS
	([])	101	¥3	(7 1)	
\$1.2	241.00		70		
302	241.11		•0	1043.00	2.110
50.5	241.31	23	30	1047 77	2.104
344	20.5*5		Cn	10	
203	242,20	33	30	1044 51	2.071
340	243.10	53.	50	1094 37	
507	243.51	34	90	1098.08	2 104
3-8	243.eu	15.	60	15 88.37	2 104
	243,90	36.	17	1999.80	2 . 24
910	244.20	34.	10	1101.00	2.194
311	244.60	38.	90	1102.41	2.173
312	244,84	•0,	20	1103.67	
213	**2.34	۰۵.	60	110	4.202
314	245.90	۶۲.	70	1106.60	4.152
31-	26 16	36.	36	3107.50	2.120
314	246.20	\$2.	76	1109.69	2.065
31,	246.50	٠.	• 0	1110.90	2.041
310	246,60	se.	90	1111.41	2.05"
14	246.40	sı.	90	1112.70	4.012
		21.	6 r	1111.56	1.051
	2		30	11:5,20	2.9-5
		ac .	30	1116,37	5.05+
375	2	30.	20	1117.53	2.022
325	288 01		00	1119,15	8.094
346	289 10		90	1151.73	2.087
317			6 9	1122,16	2.042
3.0	243 70		• 0	11.3.45	4.115
349	249.95			1124,74	2,116
3.50	256 20			1152.60	5,345
333	46.15			1750.04	2.145
-				1129,04	2.234
YU1 81	s :	. 00	1.30	57.00	
KŪ. A.	● • C:	\$1.00	180.20	100.nc	116.30

	٥Ľ	ACC FORTMULL	054-21	
L	516173462	31617124	CONDITION	COMPUTED
	CE214	16.4177. 7	DEFIN	DENSITY
	19191	10341	1541	(68/00)
1	10.00	34.90	130.64	4.033
÷.	11.00	33,97	134.59	1.935
-	11.64	43.76	13	1.471
•	12,50	19.80	139.69	1.790
•	14.94	23.60	142.03	1.995
	19.50	49.60	148.30	1.964
•		31.40	144.47	1.992
10	15.20	>1.80	151.04	2.857
42	15.80	34.61	153.54	2.7+2
13	14.75	36,80	196.92	2.077
14	17.20	90.00	138.47	1.970
1.5	10.50	41.00	163.96	1.029
4.7	14.34	21.40	167,10	1.4.7
	14.90	43.00	169.06	1.041
<i>e</i> u	24.60		172.47	1.41*
¥1	71.30	41.20	174.93	1.632
	21.40	20.00	176.10	1.013
44	23.04	22,00	181.59	3.845
43	25.43	e7.00	183.55	1.925
17	24,20	20.20	196,24	1.817
	23.24	19.00	190.20	1.794
49	13.90	ee.70	191.38	1.856
30	20.10	48.9L	143.75	1.434
30	27.34	48.90	199.93	1.951
	27.64	47.75	141,60	1.534
34	28,40	26.0C	202.73	1.920
36	29.61	28.10	.07.43	1.447
31	30.60	30.3"	200.00	1.970
3.4	N. 70	51.60	211.74	3.994
- 2	\$\$.50	43.30	/14.09	1.980
74	84.04	43,30	218.83	1,868
- 15	32.70	20.00	888.71	1.424
	54.34	39,30	\$25.06	1.803
	34.90	49.80	226.14	1.424
47	\$3.44	22.46	834,15	1.451
11	35.94	45.30	\$32.11	1.497
	37.34	26.00	236,80	3.408
÷1	38,24	25.70	241.11	1.903
	34.20	25,80	245.03	1.875
84		18,00	240.16	1.782
23		17.80	250.51	1.779
	+1.58	27.30	234.04	1.928
58	42.00	24,70	255.99	3,887
60	\$3.19	21.00	237,13	1.841
	\$3.30	23,80	261.06	1.873
	**.0*	21.70	263,61	1.840
	50	63.00	265.74	1.042
* 9	45.44	23.00	270.09	3.061
	46.00	20.70	271.66	1.424
		24,80	275.18	1.007
		29.00	279.97	1.955
	47.50	30.25	277.53	3.973
12	47.00 48.40	28.80	291.04	1.93
13		40.00	283,41	1.820
14	47.20	17.00	289,19	1.798
16	96.44	27,60	247.32	1.923
n	50.60	20.00	289.67	1.939
	50,40	26,90	298,85	1.922
	51.90	24.50	291.76	1.084
•1		19.80	297.50	1.889
	55.20	20,00	299.85	1.951
	54.20	20.20	303.77	1.017
	34.60	19,40	505.34	1.80-
*6	55.00	¥0.00	506.90	1.413
	56.69	20.00	313.17	1.013
	97.00	40.40	314.74	1.620
70	57.30	23.90	315.91	1.864
	54,10	26.00	319.04	1.908
43	54,80	23.20	\$21.78	1.864
75	57.14	24,30	328,96	1.661
76	60.00	23,50	326.40	1.868
77	60,20	17.50	327.87	1.004
	40.90	33.05	330.81	1,731
144	61.50	15.00	331.10	1.735

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	•	EACE BONEMOLE	05R-21	
J	01\$1117E0	010111260	COMPUTEN	COPPUTLE
	DEPTH. A	DEMSITE. V	24 59	DEN-11-
	(DIV)	(014)	151)	16475()
141	61.99	13.00	313.92	1 790
142	42.14	19.50	334.71	1.606
147	62.60	24,00	336.47	1.676
104	62.8V	24,38	337,45	1.641
793	65.50	23,00	340.14	1.845
106	64.00	21.00	342.15	1,029
107		10,00	343.32	1.782
1	44.60	16.00	344,30	1,751
	43.40	11.00	346,06	1.472
111	\$5.29	19.50	338.76	1.727
412	66,70	16.00	\$52.72	1.751
113	67.10	¢1.70	354,24	1.840
11.	67.70	81.90	356.64	3.045
115	68,00	23.80	257.81	1.475
444	60,40	46.20	339.30	1.911
		26,20	361.73	1.911
	30 80	24,00	362.51	1,076
340	74.20	43.VC	344	1.844
1 41	70,75	49.40	144 14	1.000
242	71.00	23.00	369.36	1.873
203	13.24	14,00	370.34	1.782
244	71.44	15,80	\$71.13	2.748
145	72.00	15.70	373,48	1.746
144	72.70	23,00	376.22	1.661
	73.00	**.00	577,34	1.676
1.49	78.40	23.10	378.96	1.865
1.30	79.54	10.20	301.7U	1.115
1 3 1	75.00	9.10	385.23	1.493
1.32	75.40	12,00	386.74	1.688
1.83	75.80	13,60	388,34	1.716
194	76,10	12,60	389.53	1.697
1 3 5	76.50	15.20	341.10	1,750
1.06	77.30	15.40	394,25	1.741
1.57	77,40	16,50	344.62	1.75*
1.49	78.40	14.80	346.47	1.724
1.48	78.79	32.30	399.12	1.491
141	79.10	15.40	101.84	1.744
148	74.50	13.60	462,65	1.723
148	79.90	15,40	+84.41	3. / 81
1	80.50	17,00	406.76	1,798
144	81.60	27.50	468,72	1.004
197	82.10	26.70	11.13	1.914
70º U	• \$:	.00 1.	50 63.40	
30. A	• e • c:	\$0.00 ZO9.	00 1.00	.25535+67

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	P	LACE BORTMULE	0#1-20				PEACE BORS	-0.E 08	1.20	
ų	DIG(7)260 DLP10. 3 TUL91	DIFITISED DENSITY - T IDIAL IVISIO	CD4#UTER DEPTH (FT)	CUMBUTEL Dénsity (6#/CC)	J.	615:112E DEPIN, N 10191	2 0161112 DENSITE 10161	-e ci	0=0UTE0 DEPT6 1FT1	(0800111 D14-11- 164/00-
1	2.00	4.70	119.82	1.917						
2	5.84	6,90	125.74	1.951	141	75.80	11,70			2.027
	*,54 *, 7 8	6.70	126.48	1,948	103		12.90		10.46	2.046
•	5.90	7.30	131.96	1 954	314	78,89 76,00	11.20		17.67	5.014
•	6.10 5 9 0	3.10	132.75	1.892	106	79.94	4.40		21.93	2,050
a	7.24	2.00	137.06	1.67	107	#C.40	8.50		23,84	1.976
	7.50	4.70	138,23	1.417	1.4	80.60 80,60	10.00		124.14 125.44	2,800
	9.00	9,60	144.11	1.997	110	81.20	9.90		27.05	1.998
13	7.30	7.54	1+5.24	1.150	111	84.30	3,50		PP0.39	1.89A 1.87%
	14.54	4.34	1.8.42	1,997	44.5	A	3,30		52.90	1.495
19	11.00	6.20	151.95		115	P3.50	7.00		135.26	1.955
17	12.24	13.00	134,91	2.079	416		7.20		38,00	1.456
	12.60	37.00	150.20	2.113	117	84.50	5.90			1.011
. U	14.00	0.00	163.70	1.464	+ + +	85.50	4.40		1.92	1.915
	14.50	*.oc	145.66	1. ***	141	85.60	10.00			2.60.
43	15.70	7.50	140,41	2.014 1.992	145	86.60	9.60		48.19	1.994
	14.30	9.54	172.72	1.992	143	84.90	9,50		149,36	1,987
	17.10	7.50	173.50	1.973	147	97,96	7.50		55.74	1.592
47	17.64	8.30	177.81	1.773	145	80.30 80.60	8.úC			1.97*
- 23	18.60	6.80	178.20	1.950	148	P9 50		i	58,77	1.857
30	10.00	10.00	100.95	2.000	129	PV.80	4.70		60.73	1.948
31	19.00	11,60	183.30	2,028	191	99.60	17,10		64.64	4.112
3 .	20.00		147.82	1,714	175	91.6U	15.00	1	67.00	2.047
34	20.50	4,90	149.17	1.920	1.04	92.00	-14,20		69.35	1.154
36	21,90	3.20	194,46	1.493	135	93.00	2.00		75 27	1.074
37	22,50 24.80	7.08	147.01	1.955	1.37	94.30	14.70		78.01	2.075
	23.00	12.19	200.54	2.033	138	**.80	13.80		80.32	2.060
40	20,90	10.00	182.50	2.000	140		12,40		83.85	2.036
-2	24.60	0.30	206.02	3,473	141	76.00	16.00		15.02	2.09=
	23.60	10.20	299.16	2.003	1*5	96,30	16.30		06.54	2.094
.,	24.80	6.30	213.86	1.94.	111	1'. SV	40.50		90.90	2.000
	26,99	8,00	214,25	1.769	1.0	98.00	10.40		92.84	2.006
	28.00	10.00	218,17	1.988	3*7	** . **	2.20		*5.9*	3.877
50	28,50	9.70	220.52	1,995	149	100.69	80		97.95	1.830
21	30.70	11,08	225,2	2.033	100	100.30			01.87	1 437
25	31.00	7.80	230.32	1.997	151	100.70	7.30		03.**	3,950
24	34.30	11.50	235,45	2.003	123	101.50	7.50		06.57	1,961
23	33.00	9.20	238.16	1.967	122	102.00	7,70	3	04.53	1.964
57	34,20	9.10	242.07	1,972	176	103.24	4,10		13,73	1
20	35.00	4.60	245,99	1.978	157	104.00	8,40		16.37	1.475
	34,20	10.00	250.49	2.014	179	104.50	7.00		14.15	1.953
1	36.50	9.90	251.87	1.998	141	103.30	1.60		23.46	1,864
- 3	38,09	9.00	255,29	2.013	3.52	104.20	5,10		24.99	1.923
	30.80	1. 10	260.88	1.910	1++	100.80	*,30		26.16	1.414
	39.30	4,30	842.44	1.721	145	107.50	1.90	5	80.08	1.474
	40,83 91,80	8.04	139.54	1.981	167	100.30	4.3D 5.8A		39.87	1.914
	41,40	9,30	273.45	1.989	100	108.60	. 70		\$4.39	1.917
/0 /1	42,40 43,00	9,90	274.09	1.990	1/0	107.00	5.60		35.96	1,931
12		0.10	\$79.30	1.975	1/1	109.70	. 60	ŝ	38.70	1.899
73	**.10	5.80	201.65	1.934	1/2	110.40	4,40	5	39,49 41,44	1.912
15	60.00	10,50	343,95	2.008	112	110.50	. 70	5	41.84	1.85
15	60.90	12.00	345.91	2.031	115	111.00	.70	5	43,80	1,854
/.	62.50	7.00	353.75	2.074	417	112.30	6.00		48.84	1.937
19	62.90	7.00	355.32	1,953	1/2	114.50	6.70		49.68	1.948
•1	63,90	13.50	336.8'	2.000	100	113.40		Š	53.20	1.984
42	64.30	29.10	360.88	2.064	147	113.60	11.00	2	53,79	2.014
	65.54	12.10	362.76		1=3	114.90	1.70	;	49.0m	2.000 1.869
	65.90	11.50	367.87	· · ·	144	113.20	1.*0	5	60.26	1.866
	67.00	12.50	\$69.43	2.03*	1*6	116.14			67.61 63.78	1.915
	67.44	12.60	372.95	2.041	187	116.40	6.00	5	64 92	1.937
99 70	68.90	12.80	375.69	2.020	744	117.00	6.70	5	47.31	1.948
21	\$9.30	13.20	379.61	2,050	3 7 0	117.70	7.90	ś	70.05	1.967
42	64.70 70.70	11.A0	361.96	2.028	171	110.00	8.00	3	71.23	1.969
	11.20	11.70	387.84	2.047				,	4	
**	72.10	10.30	391.37	2.005						
77	73.64	13.80	\$97.25	2.091	10	3 1	10.00	2.00	63.60	
78	75.00	16,30	399.21	2.099	¥0	8. C:	30.00 2	26.90	258	
100	75.50	13,00	404.49	2.057			•••••			

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	•	SACE PORTHOLL	001-14						
J	016111200	01-117280	COMPUTER	COMPUTED			LELL BORINGL	1 001+14	
	40191	DENSITY. I IDIVI	DEPTH	DENSITE IGN/CC)	5	DEPIN, I (D1y)	DINITIZIO DENSITI I IDINI	41704003	678-0116 059517+
ź	29.70	35.00	238.57	2.057	1.1	77.90	35.60	*** **	
3	30.60	10.70	242.50	1.903	105	78,40	34.00	\$32.73	2.0**
;	30.50	30.60	244.47	1.981	1.04	79.50	33.40	*35.P*	2.025
•	91.50	44.00	248.40	1.972	142	80,50	35.20	480.44	2.052
	52.0°	48.50	250,36	1.948	106	61.00	29.65	442.95	1.969
Ţ	52.90	47,80	251.93	1.969	146	82.50	47.60	***.10	1.965
1.0	33.20	42.00	255.08	2.083	1.44	84.00	33.20	150.81	2.022
11	35,70	33.20	257.04	4.022	110	#3,70 #9,30	\$4,00	453.57	2.035
13	34.50	35,68	240.19	2,053	112	84,99	34.00	*35.92	2.064
14	35,00	33.10	262.15	4.057	115	45.20	35.20	455.46	2.05
14	56.00	27.30	264.12	1,981	115	85.9u	31,30	461.03	L.995
17	36.60	47.80	768.44	1.975	146	46.04	48.00	*62.61	2.005
	37.0U	41.40	271,59	2.000	118	47.14	41.20	\$63,75	1.991
20	50,00	\$4.30	273.94	4.039	119	87.64	56,70	168.81	2,016
	30,50	34.10	275.91	2.036	140	86.1V 86.70	35.80	470 84	2.063
23	39.40	33.60	279.45	2.035	144	47.00	47.30	473,22 474 MD	1,956
	39,70	32.90	280.63	2.017	143	90.00	32.00	\$78.33	2.00
40	40.40	20,50	282.92	1.948	145	\$1.00	31.00	479.90	1,987
47	41.00	30.30	203.74	1.976	14	91.70	39.00	445.01	2.047
	41.70	\$4.00	285.44	2.035	120	42,00 97,10	48.10	***.3*	2.102
30		35.30	290.06	2.028	149	94.60	43,70	\$47,37	2.050
97	* 3, *0	\$3.30	295.17	2.024	190	93.00	30.90	**0,12	1.984
35	43.80	34.50	296.74	2.042	1.32	99.00	33.60	**1.69	2.028
34		37.40	565.67	2.082	195	** . 54	36.00	76.01	2.035
33	45.00	34,70	301.46	2.046	145	99,90	\$3.00	497.59	8.050
\$7	*6.00	32.08	305.39	2.019	136	93.60	47.40	498.37	2.082
30	**.**	45.00	306.76	2.050	137	94.50	46.70	503.09	2.048
	N7. BU	45.40 49.40	509.32	2.057	1.07	47.20	34.30	304.66	2.039
21	47,50	43.20	311.20	2.933	140	\$7.50	\$5.00	367.81	2.021
	44.CU 54.50	\$2.50	\$13,25	2.011	145	47.80		308.98	2.035
		34.10	315,21	2.017	1 * 5	40,70	35.60	507.30	2.050
23		37.90	317.94	2.096	144	77.DU	35.00	13,70	2.050
	59.54	47.10	319,93	2.083	1.4	100.00	32.60	515,27	1.997
9	51,00	\$3.00	325.01	2.018	147	100.30	\$2.00	518.81	2.011
	51,30	33.90	326.22	2.064	144	100.80	\$\$,30	520.70	2.024
21	52.30	38.40	328.18	2.080	100	101.60	27.00	522.74	1.991
23	52,70	.4.20	331.72	2.030	121	102.00	e6.50	525,44	1.927
34	55.20	11.40	333.69	1.994	122	102.30	27,90	326.67	1.939
23	54.20	\$2.6P	337.62	2.013	124	103.00	49.10	529.82	1.921
26	54.70	\$0.00	339,50	1.972	155	103.80	48.00	532 57	1.940
58	55,70	<7.30 Ch 00	340,76	1.961	157	104.40	29.30	533,35	3.961
	54.50	49.70	345,47	1.967	128	104.90	20.30	14	1.76
60 61	57.00	30.30	348,62	1.976	100	105.80	e8.70	339,64	3.951
• 2	54,50	\$4.70	393.74	2.011	1 . 1	104,60	31.80	541.71	1.987
**	59.00	94.00	346.48	4.035	162	107,30	29.20	346.32	1,939
+5	60.30	29.50	359.23	1.980	164	108.20	51.00 K9.00	547.90	3.967
	60.60	69.90	362.77	1.970	165	104,90	28.70	352.41	1,956
	62.8V	38.09	366.31	1.689	167	109.40	47.80	354,54	1.937
69	62.50	20.00	870.24	1.877	2.6.0	110.00	10.00	336.54	1.987
10	63.20	×6.00	372.99	1,909	170	110,70	32.50	559.69	2.013
12		27.90	377.71	1.030	1/1	\$\$1.90	33.00	540.87	2.003
	67,90 63.60	33.50	379.67	2.027	1/2	112 10	43.5n	563.23	2.027
75	65.60	49.89 91.40	380.07	2.126	1/4	112.00	32.00	565.19	2.016
1	63.74	40.00	302.02	2.127	175	113.00	32.90	366.73	2.017
7.	64.70	35.30	345,18	2.118	117	119.50	13.50	572.66	2.024
19	67.20	91.30	388.75	2.129	1/6	115.00	34,10	573.84	2.034
•0	\$7.5U	41.20	549.49	2.140	1 19	115.60	53.0C	579.73	2.01*
	4	44,30	393,0%	2.13	1*1	110.90	34.40	581.31	2.041
	70.00	38.40	599.72	2.104/	1#2	117.20	34.40	185.24	2.030
	70,70	48,00 35.00	400.90	2.997	103	117.6V 318.60	32.00	586.81	2.003
*	73.0U	\$2.00	402.47	2.003	1=5	114,60	33.40	568.38	2.050
87	71.30	\$2.00		2.003	146	117.10	36.20	592.76	2.069
.,	72.00	<#.10 \$2.08	*06.79	1.942	100	17.80	54.20	595.46	2.038
*0	72.50	\$2.40	409.54	2.003	14.4	120,90	33.20	597.81 599.74	2.031
71	73,00	\$3,70	111.51	2.030	141	121.50	\$3.60	602.14	2.020
*5	74.00	30,00	*14,26	1.972	172	122,40	49,00	604.10	1.936
74	74.50	31.20	14.67	1.991	173	122.80	\$2.00	603.67	2.003
76	75.20	34,40 18 Au	419.37	2.041	344	122.90	\$1.00	607.64	1.987
*7	76.00	33.40	*26.14	2.024	1 76	124.00	20.00	\$10.00	2.050
78	76.69	\$3,00	475.44	2.019	177	124.50	35.70	613.14	«.035 2.061
140	77.20	33.49	*2*,23	2.025	7.4.4	125,30	#5,06 38 8c	\$15.11	2.050
	-		- 20. 12	<.033	200	120.00	10,60	617.07	2.947
							-	******	••••

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		PLACE BUREHOLE	091-19				ATT BORTON		
J	01617575	0 016111250	(DAPUTI 5	COMPETEL					
	10141	0E451774 7 40149	0[PT++ (FT)	15N52** 16#/CC1	•	{ ! # ! 4	UENSITE -	(P1	
	126 68								
e4e	121.70	35,00	672.97	4,019	1		*0.00	547.66	.184
203	120.00	*****	671,68	2.0.4	5	47.54	28,30	147.52	2.155
205	179.00	\$2.30	631 63	2.014	•	48.40	\$7.00	102.61	4.132
246	129.90	25,00	413.58	2.090		49.00 89.50	41.00	401.25	2.050
400	130 40	34,76	637.12	2.046	Ţ	50.60	40.40	427,39	5.018
204	151.40	\$3.50		2.027	4	·1.0"	41.40	\$15,45	2.037
<10 <11	131.60	\$3,10	\$42.62	2.020	10	51.70	31,30 32,80	415.97	2.035
412	112.70	33.80	*** **	2.028	41	53.24	45,20	175.77	2.105
e 1.5	150.00	35.00	417.34	2.050	12	54.00	35.20	• 26 . 10	4.102
413	133,20	34,5j 34,20	648,12	2.039	14	55.50	10,90	411.14	2.021
136	194,10	43. 40		2.025	15	N6.20	10.10	* 36.15	2.611
217	134.30	\$1.8C	617.64	1.987	17	59.00	34.50	439.56	2.090
419	195.20	30,90	611.98	2,006	18	54.70	37.70	**1.17	
440	152,70	33.00	697.44	2.01*	14	60.20 61.04	39.00	44.4.31	
447	136.40	32.00	661.09	1.986	4 1	61.50	*1,10		20
443	137.00	32.00	\$65.06	2.003	~~~~~~~~~~~~~~~~~~~~~~~~~~~~~~~~~~~~~~~	62.00	*0.0-	461.04	2.104
***	130.30	36,30	566.20	2.071		63.30	35.70	466.62	2.076
***	138.50	38.30	468.45	2.102	45	63.70	\$5.00	458.34	2.044
448	130.90	38.00	670.23	2.047	21	62.30	33,80	4 1 7 7	2.010
469	1.0.10	34.30	675.24	2.050	6 8	66.70	23.70	481.71	2.076
4 5 7	140.80	35.30	\$77.99	2.055	29	5°.70	36.00	*#5.50	2.11*
232	142.10	34.90	679.94	2.035	31	68.6U	15.00	469.57	2.073
233	142.40		6A4,28	1.972	32	69.20	34.00	**1.**	2.081
234	141.90	30.40	686.25	1.978	3,	70.40	30,00	415.23	4.150
236	1+3.54	34.50	488.51	2.042	35	12.00	47.90	503.96	2.140
237	143.40	33.00	698.10	2.031	46	72.70	47.30	306,96	2.110
249	145.64	34,30	692.93	2.050		74.74	\$7.80	515.55	4.144
240	145.90	\$3.90	\$98.04	7.027	39	75.20	36.00	517.69	5.11.
	146.70	43.90	498,83 781 18	2.844	•	78.70	44.40	371.98	2.093
	1=7.00	32.00	702.34	2.003	11	79.30	33.00	536.15	2.064
	146.20	31,00	784,33	1.987		80.00	19.00	558,29	1.825
£*6	149.00	34,50	710.22	2.042	• •	81.00	19.00	342.56	1.825
478	188.09	44.80 Al 50	712.93	2.035	47	82.40	44,60 90,00	544.73	2.091
249	150,70	29.20	716.91	1.957	<u>4</u>	80.30	47.30	332.44	7.130
230	191.70	15.00 36 30	720.84	2.050	50	83,76	38,10	554.17	2.131
1.55	154.60	46.20	724.37	2.069	21	85.00	32.00	154.75	2.0+1
235	153.00	35,40	725.95	2.057	22	-5.jv	28.50	560.38	1,987
275	154.04	33,00	729,88	2,200	34	86.75	39,00	361.90	2.167
276	135.00	\$4,20	733.41	2.034	35	87.30	40,00	569.42	2.184
238	196.20	39,60	737.34	2.^60	51	90.00	57,50	5 . 5	2.141
437	197.00	35.70	741.67	2.061	24	91.00	\$7.70	505.50	4.1.
441	130.00	34,30	744.42	2.039		92.50	>3.50 A5 80	787.44	4.075
462	157.00	36.40	749.53	2.072	+1	92.8U	38,30	5	2.13
200	157.50	35,80	751.49	2.063	• 3	94,00	*1.80	596.30	2.201
285	160.50	\$4.50	755.42	2.014		95.60	34,70	603.25	4.150
266	160.90	35.10	757.80	2.052	• 5	76.00	20,20	605.96	1.987
448	162.00	#2,50 #3,00	759.36	5.011	.,	97.50	26,50	604.44	1,953
249	162.40	32.00	762.89	2.093	68	90.00	46.91	413.55	1.944
611	165.60	20.43 20.43	766.Du	1.947	/0	94.00	24,70	618.4ª	
					(1	99.60	46.20	4.53,27	2.119
					12	101.00	44.50	624,62	2.000
10. 4.	51	.00 1.90	63.60			101.00		**:*	4.150
	a. (:	30.00			/5	102.20	46,90	613.57	2.124
-	2		1.00	. * 34 4 2 5 3 9	17	183.84	e9.50	697,00	2.00-
						104.00	-1.00	6×1.*C	2.201
						106.30	20.90	647.51 631 17	1.960
					* 1	107.00	26.00	65 7	1.944
					÷1	108.60	20,0C 26,30	61 - 69 661 - 64	1.979
						184.70	25,00	\$61.47	1.927
					•1	107.20	26.00	443.41	1
						110.50	\$2,60	667.17	2.05
					**	111.00	29.50	676.77	
					vo	118.00	11.30	677.35	7.056
					21	114.00	7.60	483.34	1.630
					73	115.60	39.60	696,51 691 04	1.900
					**	116.80	#1.00		2.030
					¥6	110.50	##,30	699,24 783	4.172
					**	119.00	48.45	703.67	2.071
					74	117.30	40.00	786.96	2.013
					100	120.20	30.00	710.02	2.071 2.013

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		E-CE BOREHOLE	082-04				ALL BOALHING		
•	DISTICO DLAID- E (Uly)	DIVITIZEC DENSITE T (Civ)	2049UT[r DEPTH (Ft)	CUMPUTEC DENGITT 168/CC1	L	DISITIZED DEP34. 2 TUIVI	D14111210 D14111210 D14111210	(DROUTE- OCOTE- (1 T 1	COMPUTEC DENSIII IMR/LCI
101	120.60	42.30	712.54	4,052					
103	121.00	46,50	714,26	1.953	445	210.50	33.70	1048.38	4.06*
104	122,30	44.30	719.04	1.984	2L3	210.00	35.20	1049.67	4.102
102	122.80	34.71	723,44	2.043	445	211.60	46.5n	1102.67	2.103
147	124.04	31.76	727.15	2.025	206	213.00	36.40	1109.11	2.122
108	125.60	33,70	731.41	2.076	eu .	214.00	31.30	1110.40	2.075
110	126.40	27.50	737.43	1.970	209	214.60	92.50	1115.97	2.056
111	126,90	29,70	789.54	2.008	211	216.30	41.20	1120.70	2.036
113	127.80	\$2.00	745.44	2.047	412	217.30	34.20	1127,36	2.065
110	127.00	35,50	748.55	2.107	414	217.00	39.81	1150.07	2.180
114	130.40	37.60	754,60	2.143	415	220.00	55.00 52.85	1139,14	2.440
114	191.50	43.00	759.32	1.893	217	221.40	22.50	1145,16	2.547
114	132.50	14.67	763.61	1.750	21 9	224.00	39.00	1151.60	2,167
141	134.45	a. 50	770.0	2.175	220	225.20	44.70	1161.47	2.264
145	136,50	38.20	779.92	2.153	242	228.50	•0.ce	1169,14	2.926
144	137.90	36.50	795.37	2.124	223	230.09	34.70	1182.07	2.04*
146	141.10	41.20	800.52	2.204	243	233,30	36.00	1197.09	2.115
247	1+3.10	29.00		2.167	246	234.CU 234.4U	34,00	1199.25	2.001
149	345,34	38.90	918,12	2.127	248	239.00	\$7.00	1203.73	4.134
1.00	344.50	\$2.50	823.70	2.056	664 630	235.44	41.50	1205.25	2.209
125	147,84	32.00	N79.28	4.047	431	237.50	23,50	1214.26	2.415
133	148,20	51.00	831.00	2.030	232	230,50	35.40	1217,69 1222 Au	2,415
125	150,70	\$3.30	841,73	2,10	254	240.50	36,8C	1227,13	2.129
136	191.13	32.20	843.44 847 31	2.050	< 36	241.80	36.20	1229,28	2.197
1.38	152.40	35.00	849.02	2.094	457	242.00	25.10	1253.57	4.100
137	155,00	33.0C #8.20	851.60	2.0%8	265	245.44	25.8P	1237,43	1.941
1-1	154.40	42.20	\$57.61	4.050	240	244,10	26.10	1242,58	1.946
1 . 3	130.00	10.00	844.47	2.150	242	245,40	45.27	17+4.14	1,431
	134.40	87.00 37.90	866.19	2,132	211	216.00	45.80	1249.45	1,941
1 **	197.59	36.00	870.41	8,115	243	246.20	45.80	1251.60	1.941
194	130.50	37.50	\$75.20	2.115	247	246,70	24.40	1252.03	1,917
199	159.20	38.10	878.21	2.151	245	266,90 267,30	23,30	1254,60	1.894
121	160.50	38.30	882.93	4.135	238	247.50	15.20	1257.51	1.897
722	161.80	31.90 32.80	849,37 891 55	2.045	492	248,40	29.30	1258.03	1.917
1.84	163.10			2.129	253	248.70	26,30	1262.33	1.950
124	144.24	\$6.30	908.25	2.121	(25	238.10	45.20	1268.34	2.273
127	166.90	17.0C	11,25	8.132	437	250,30	*3.50	1269.19	2,288
139	157.00		715,12	8,115	258 201	250.90		1271.77	2.278
141	160.90	#3.00 #7.30	914.83 918.12	2.098	260	252.00	42.70	1276.34	2.131
1.55	149.00	30.50	980.27	2.158	(*) (*)	232.24	43,20	1277.35	2.060
1	171,50	36.80	930.14	2.261	24.3	252.40	\$5.10	1279.92	8.100
192	174.50	34,80	943.87	2.095	244	253,84 254,70	45.40	1284.22	2.276
247	176.00	33.00	950.31	2.044	266	255,00	44.10	1289.57	2.083
169	177.00	33.00 35.00	934.60 961 Ob	2.06% 2.09%	268	255,64	33.70	1290,65	2.076
1/0	179.40	-0.00	944.90	2.184	469	256.20	33.20	1294.52	2.966
112	101.20	**.85	972.63	2.252	413	257.40	50.90	1299.67	2.003
1/5	183.80	39,00	,,,,	2.167	212	258,79	44,30 58,50	1305.25	2.046
115	144.64	34,60	987.22	2.075	e**	259,64	32.90	1309,11	2.040
117	184.30	32.50	994.52	2.054	215	260.10	29.10	1311,25	1.997
1/4	107.60	\$7.30	1000.09	4.138	217	260.50	29.10	1312,97	1.99
100	189.00	35.00 \$7.50	1004,34	2.098	419	260.80	\$0.30 (9.10	1314,76	2,010
191	140.00	38.00	1010.40	2.150	4#C	261.10	28.80	1315,55	1.997
742	191.14	34,20	1012.11	2.170	442	262.00	49,00 87.60	1316.40	1,996
104	191.70	35,80	1917.67	4.112	e*3	262.30	49.30	1 \$20.70	2.004
144	148.50	 	1023.42	2.126 2.008	243	263.00	\$2.10	1325.70	2.038
187	194.60	34,00	1030	2.001	286	263.80	41.90	1327.13	2.845
187	1	\$5,50	1034.00	2.107	298	264.60	40.10	1330.51	2.025
140	197.60	40,00 Al.80	1043.01	2.165	249 270	263.30 263.40	30.20	1333.57	2.016
172	199.50	29.30	1031.17	2.001	271	266.00	29.20	2336.50	1.999
7.84	500.00 500.00	38.30 48,30	1053.31	2.198	543	266.30	29,00	1337,43	1.995
1 7 5	203.30		1058.84	2.105	294 194	266,60	. 70	1339.15	1.991
141	.03.00	21,00	1064.19	2.030	c76	267.50	29.30	1348.01	2,801
144	206,00 207.60	33.60 32.70	1079.06	8.109	477 478	267.90 268.80	28.90 29.40	1344.73	1,994
440	209.00	34,50	1091.94	2.090	299	268.60	29.40	346.88	2.003
					340	204. 9 V	e7,99	1348.59	5.017

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		PEACE BORE	-01E 092-04			,	Start Boutun		
J	016111760 DCP14+ x))	0 00000tr	COMPUTES	ں۔ ا	DIFITIZED	216111241	COmpleter	e jime i t
	19141	10141	11.	100/001		7[#14, 3 (cly)	DENSANY. PLINA	SER HA	ng sa ta Sig∎ ta
301	269								
542	267.36	\$2.50	1341.60	8.036	į	3.DL	<3.6C	182.4	4.91
303	270.40	35,20	1353, 31	2.060	•	5.00	29.1	186.34	• • 1 1 •
545	278.60		1936. 12	2.162	•	5.20	38.1	191.7+	2.10
304	270.00	19.00	15-7.18	2.180		2.70	•0.3	143.75	2.1
• •	271.50	34,70	1367.18	4.170	,		43. Ar	1	2.18
	472.41	:1.01	1340.61	2.146	•	1.31	16.60	144.31	2.15
341	272.00	38.60	1361.47	2.160	10	8.00	*1.CC	200.12	2.14
313	272.20	40.60	1363.18	2.162	41	0.20	*1.0c	202.87	2.1**
313	272.50	*3,40	1364.47	2.240	12	6,70 9 60	*1.16	204.84	2.34
215	274 00	·1.50	3367 48	2.209	1.	10.50	.1.50	224,19	2.746
316	274,50	\$2.60	1373.06	2.057	15	11.19	38.50	214.64	2.104
318	275.50	\$2.00 \$2.80	1574,77	2.047	17	14.50	34.25	217.8	1.991
51.0	276.50	34.00	1380.78	2.047	18	12.00	35.20	721. **	1.4
341	276.60	\$1.40	15*1.21	4.037		14.40	22.5	224	2.00
342	277.00	49.20	1781.64	4.017	<1 ×1	17.11	19 1		1.12
343	/77.50	48.10	1 *85.9*	1.941	~ ~ ~ ~ ~ ~ ~ ~ ~ ~ ~ ~ ~ ~ ~ ~ ~ ~ ~ ~	13,59	56,01	2 . 1 . 4 3	2.076
343	278.10	67.31	1346,74	1.98		16.4.	36.		2.092
	274.50	27.30	130		<i>c</i> .	1	41.H.	2.0.0	2.187
328	279.00	30.70	1397.2	4.72	42	10	44 70	234	2.191
329	279.20	\$2.00	13-3.21	.032	< 0	18.50		242.6	2.715
330	279.70	31.60	1395, 37	2.040	24	10.60	39.00	242.42	2.112
332	280.60	41.60	1399.24	2.032	31	14,30	«C.e.	246.51	2,018
393	281.00	49.60	1400.95	2.000		17.65	17.30	248.54	1.772
225	281.8U 281.8U	30,30	1404,34	2.018		24.50	23.00 20.00	252.44	1,841
336	204.30	42,50	1405.67	2.036	2.4	24.00	41.00	2 1 20	1.98
337	282,40	51.80	1406,96	2.0++	21	72.56	\$4.00	258.54	2.03.
1	283.50	30.40	1404.54	2.045	56	23.50	37.61	241.11	2.075
371	208.60	** .*0	1412.11	2.000	9 E 9 D	23,65	26.31	26.5	2.270
3.92	284,24	e9.80	1412.97	5.010	•1	24.60		266.65	2.191
** 5	284,90	30.00	1417.69	4.015		25.00	1.50	269.01	2.151
	265.20	30.40	1416.55	2.020	••	26.20	34.00	272.16	2.034
	\$85,50	40.40	3+20.27	a	• •	26.50	\$3.0C	274 42	2.018
	246.00	29.90	3488.43	2.011	.,	21.60	\$2.30	\$79.25	2.007
344	286.50	31.80	1 4 2 4 . 5 6	2.044	*8	26.60	35.30	283.1*	2.034
370	200,00	\$1.50	1.24.24	2.037	• •	29.01	\$7.80	28	2.084
372	288.24	31,40	1435.82	2,038	21	29,90	36,00	286.73	2.048
339	204,50	10.30	1***.1*	2.01#	21	30.30	31.30	284.88	4.52*
275		41.40	1434.86	2.011	5.	31.00	33,00	243.61	≪."][2.018
335	287.74	35.30	1.38.29	2 047	**	31.40	\$1.40	294.43	1.993
354	294.30	33.40	1440.01	4.071	.,	34.50	33,00 30,00	246.47	
379	540.00	32,40	1443.01	2.054	58	35.00	33.60	300 1	
301	291.30	34,10	1445.16	51083	-0	33.80	36.70	303.66	2.076
362	292.20	46.00	1449.02	8.115	•1	54.70	39.20	307.20	2.967
36.4	292,50	35.30	1+50,51	2.103	• 2	35.00	38.00	508.39	2.191
363	293.00	39.90	1912.94	2.115	••		36.30	512.14	2.070
366	295.40 295.40	44.90	1454.17	8.165	**	36.60	\$7.40	514.48	2.087
358	299,10	36.80	1456,32	2.165	67	37.84	44.00	316.26	2.045
314	294,30	45.00	1.58.05	2.098	5.5 	37.70	34.80	319.01	2.046
\$ 1	295.10	36.50	1454,32	2.121	/0	50.45	41.90 40.70	320,94	2.156
3/2	295.20	47.20	1461.90	2.134	11	39.10	•0.70	524.53	2.15
514	296.34	29.7D	1969,90	2.030	75	40.10	#3.40 \$5.00	326.89	2.024
\$13	296.40	49.50	1467.05	2.00-			36.80	351.22	2.079
377	298.80 297.20	43,80 35 mm	1468.76	2.07a	13	41,30 42,80	*0.90	533.24	2.1.2
5/8	297.80	\$5.70	1470.48	2.105		42.20	39,40	335 94	2 134
334	298.20	15.70	1.74.77	2.110	14	42,40	34,40	334	2 . 1 1 .
301	299.50	36,90	1976.06	2.131	*0	43.60	40.0r	339.48	2.126
302	297.80	\$2.30	1*#1.4*	2,052	•1	45.80	59.00	343.03	2.041
364	300.20	32.00	1+83.34	2.05*		44.5U	17.40	3 0 0 0	2.08
	301.30	10.40	1988.04	2.030		45.50	46.50	347,36	×.177
346	302.50	44.30	1443.25	2,121	*5	65.AU	.0.76	355.90	2.139
	40	48.30	3994,94	2.083	• 7 • 7	\$7.70	56.70	355.63	2.076
304	303.90	2.60	1	8.037			40,70	361.14	2.139
AV1	304,18	33,90	1900.09	8.079	**	47.ju	39.80	363.90	2.12
3.55	303.20	34.80	1503,10	2.075	•1	50.50	34,10	366,63	2.057
	505,60	34.90	1506.53	2.047	72	50.60	\$5.20	369,80	2.033
379	326.89	JU. JU JU. JU	1308,25	2.131	**	54.00	24.10 35.00	370.98	2.11
376	306.94	49.70	1918.11	8.179	*5	94.70	16. Jr	576.0	5.674
	384,04	#3.30 35.10	1514.69	2.107	**	58.60	32,70	\$\$1,61	2.013
379	308,90		1519.54	2.146	78	56.00	34.00	383.19	2.05
					100	56.90	44.20	394.61	2.1**
							46,TN	346.27	1.908
101 81	S :	.00 1	.50 58.85						

X-- A- H- C: 30.00 323.70 570.00 114.50

		TACL BURFINGE	-01-04			F	FACE BOREHOLE	001-05	
J	2131114Er	1611111 (hette	THPUTCO	(DRAUTEC	J	016111260	016111740	COMPUTES	C0#PU1EC
	10141	le ly i	1811	1987001		0[PTH. 4	DENSITT T	DEPTH	CENSLEY
							(019)	(***)	(GR/SC)
141	58.00	22.00	396.94	1.845	202	115.00	¥8.00		
143	37.50	#7,30 #8,64	402.04	2.086	142	119.00	47.00	620,90	2,238
194		47.00	406.81	2.081	203	112.10	48.90 99.40	623,74	2.260
148	62.14	94.00 40.00	408.78 711.14	2.112	445	17.30	\$7,00	632.40	2.230
147	62.20	46,60	+15.47	2.675	207	31.70	96,60 99,00	433,9# 437 41	2,282
744	64.74	38,80	*17.41	7.283	208	119.20	*8.00	639.84	2,254
110	56.04			2.3+0	210	129.00	*9.00 ##.10	641,46	2.269
	4/.30	47,40	+39.55	2.274	441	170.20	49,74	444.42	2.175
444	\$8.09	51.90	438.51	8.305	\$13	121.40	98.40 97.50	646.18	2,103
		.7.		2.249	234	171.00	\$1.00	610.12	2.501
314	78.00	NE.80	446.18	2.266	216	173.30	45.40	456.02	2.213
110	71.24	48.20		2.257	417	124.70	\$6.20	657.60	2.225
140	73,00 74.60	35,30	453.27	2.337	213	126.00	A. 10	\$65.65	2.274
14	73.00	54.50	*57,93	2.336	240	127.00	\$2.70	670.59	2.327
248	73.30 73.84	35,30	*57.17	2.368	242	170.00	\$5.70	674.55	2,343
144	74.34	57.40	\$63.11	2.001	223	150.50	P1.50	676.90	2.306
1.4	75.00	39,31	465.47	2.032	243	129.50	1.00	680.43	2.301
107	76.00	· 4.0r		2.426	226	190.00	22.80	6.2.40	2.329
148	76.51	50,21	*71.77	2.414	228	131.00	31.00	686,34	2.301
1.5		>6.50	.75.71	2.347	229	131,90	50.05	689.80	2,285
121	78,24	51.00	478.46 878 38	2.301	231	133,00	1,00	694,21	2,301
1 3 5	78.60			2.277	232	133.20	>1.70	693.00	2.312
192	79.90	30.30	482.79	2.290	234	135.00	31.40	702.09	2,307
1.86	80.50	21.00	487.52	8.301	235	155.00	2.00	703.66	2.316
137	80.80 Al.40	50.00	488,70	2,285	237	137.04	99.20	704.44	2.272
134	\$1.90	92.30	**3.03	2.164	236	157,90	•0.30	713.50	2.250
140	A4,00	**.**	447.36 501.30	2.265	470	139.00	\$0.50	717.63	2,273
142	84.40	50.00	502.07	2.245	243	139,3v 140,0v	>1.20	719.01	2.304
144	43.20		506.02	2.272	e = 5	1.0.00	\$1.70	723.34	2,512
145	85. 6D	-7.00	507.60	2.210	201 213	140.90	70.80	725,31	2.297
1.17	#4.50	25.80	511.14	2,329	414 	141.90	**.70	729,25	2.282
148	85.7V	91,00 Su.47	511.43	8.501	c	143.40	**.30	730.83	2,230
170	87.80	4.20	516.26	2.225	249	544.50	47.20	758.70	2.211
123	40,20 80,60	49.50 30.6P	517.85	2,271	<>)	145.50	•7.00	743.45	2.234
193	47.24	**.30	321.77	8.250	432	1=6.20	50.10	746.18	2.286
1.05	87.94	*5.60	522.56	8.885	<i>(</i>) ,	1.,,00	44.60	749.35	2.254
176	48,39	*5.80	525,71	8.885	435	147. <i>80</i>	12.70	752.46	2.170
170	91.00		528.07	2.230	237	1	*6.80	757 20	2,255
179	91,40	*5.10	330.41	2.204	238 239	198,70	90,50 48,70	759.94	2.273
4+1	92.50	**.20	554.76	2,272	460	150.80	50.00	764.29	2.285
1+2	92.80	*8,90	535.7.	2.254	 4 b 2	191.30	**.00	766,26	2.222
1.64	93.50	*5.50	538.70	2.21	263	192.40	**.00	770.54	2.234
145	94.3U 94.6U	*3.00	541,04	2,206	2×3	153.50	73.00	772.95	2.548
167	95.00	*3.00	344.61	2.175	206	153.90	35.00	776.50	2.363
1	96.30	12.80	346,97	2,199	***	154.60	**.00	778,07	2.332 2.267
1/0	96.70	**.80	551.31	2.20	609 470	155.50	47.00	182.01	2.238
1.72		48,70	335.66	2.245	211	154.40	*6.30	786.34	2.227
1/3	95,4U 95,4U	16.80	557.99	2.235	212	157.30	45.20	787.88	2.210
115	. 50	52.50	562.32	2.921	614	158.00		792.64	2.195
2/5	100.00	31.40	564.24	2.307	215	150,70	49.DO 47.40	743.39	2.269
1.10	100.64	-5.00	567.44	2.214	417	159.60		199,94	7.268
2/9	101.50	>1.00	570.20	2.301	413	160.60	·3.00	800.51	2.175
1#1	104.70	>1.00	574.92	2,901	4.0	161.30	•3.30	\$05.65	2.180
144	103.00	26.60	577.64	2.579	201	162,20	42.70 43.40	809.17	2.170
1.8.9	104.84	47,40	543,19	2.74	203	163.40	·3.20	013.90	2.178
144	105.70 106.0V	54,00	586,73	2.348	244	164,30	**. 70	#13,08 #17 **	2.202
107	106.64	57.30	590,27	3.000	496	169.00		828,28	2,260
1	107.00	22.20 25.80	591.05	2,319	244	163.60	47.60	822,56 821 14	2,250
1 90	100.00	52.30	595,79	2,321	209	166.20	91.20	824 32	2.304
192	108.50	23,60	597,76	2,341	471	167.50	33,20	823,71 825 94	2.535
293	109.20	51.70	600.51	2.314	292	167.70	-5.30	\$30.83	2.211
194	109,74	53.10	602.44	2.334	273 274	160,00 160,80	44.4C 45.20	832,01	2.147
1 76	111,10	49.20	607.99	2.272	275	169.00		835.94	2,733
177	112,00	\$2.30	411.53	2.321	276	167,60 170,0v	46.10 47.10	839,09	2,224
2 49	113.54	46.00	616.69	2.25+	(78	170 60	16.30	842.24	2.227
200	119.90	*3.00	\$17.83	2.204	240	171.50	48.40 46.20	843.82	2.260
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	•	EASE BOHEHOLE	061-95			0	CALL BONTHOLD	Chd. /	
ي	DIGITIZIO Dept-1, 1 10101	01+11725 DENSITE T (CLV)	(D4007)- J7 рун (F1)	(UMPUTEC DENSIT* (GM/CC)	J.	21111200 1-9144 8 10141	016111262 06451774 - 4 10141	1340(***- (101) (F1)	-0#06110 -194-11 194/221
\$U]	176.80	*7.60	548.9 4	2.247	1	5,50	20.3D	1=1.87	1. 4 . P
56.9	173.40	28.50	893.27	2.293	5	2,10	24,70	1	1.970
304	374.00	46.80		2,235	•	7,40	24.50	147.64	1.804
346	176.00		863.90	2.230	, ,	9.00	35.70	173.71	
Su 7	176.20	48,17	644,29	2.255	,	14.06	90.00	1.4.4.4.6	1.974
308	177 20	47,9C	866.25	2.227)0.au	57.31.	167.4-	2.091
310			970.20	2.204	10	11,00	40.00	164.75	
541	170.00			2.215	* *	14.60	44.8C	167.4	1.924
212		2.46	073.31	4.172	12	15.00	30,00	121 0	1.993
21.6	174.40	45.80	878.46	2.219		14.70	\$5.20	17.15	4.02
145	180,30	\$9.50	880.43	2.277	12	14.00	35. 1	1 3.3	2.0-9
317	181.80	38.90	884.34	8.111	10	14.30	30.00		1.71
	184.50	•3.50	889.09	2,183	10	15.20	\$1.BC	1.14.52	
219 340	183,20	41.40	**1 .*5	2.151	17	15.70	27.00	174.44	1.80
343	184.50	*3.30	846.47	4.180	41	18.50	34.00	143.1	1.78*
342	185,20	42.90	899,72	2.173	42	17.00	18.50	1	3 - 79 -
344	187.00	43,70	906.81	2,185	~ · ·	17.60	30.70	1.47	1.984
345	147.40	*3,20	909,96	2.170	< 5	10.00	30.2	1.0.4	1. 774
348	100.50	42.20	912.72	8.254	4	18.20	31.6C		2.001
348	40.00		418.62	2.279	4	19.40	\$1.00	193.	e. C
349	140.50	47,90	720.54	2.252	~	19.90	27.00	101.16	1.927
300	191.00	40.30	926.10	2.133	30	20,10	32.80	19 3	2.014
332	192.20	*3.00	\$27.28	2.175	24	71.60	\$2.00		7.006
333	192.60	•1.70	928,85	5.122		21.40	52.00	212.44	2.006
335	194.50	39.60	936.34	4.182	37	22.60	55.50	206.75	1030
336	196.20	*1.70	943.03	8.195	36	23.20	\$2.80	204.00	5.414
337	196.60	*1.80	944.61 987 74	2.15	57	25.50	34.10	210.2.	2.040
339	197.00	21.20	949.35	2.30		25.10	45.80	216.46	2.064
340	194.30	53.20	951.30	8.335	•0	26.00	33.07	214,46	4.031
342	199.30	21,90	955.24	2.277		21.54	77,75	225 53	1 1
2 4 3			446.47	2.191	••	4.50	39.5'	2.0.1	
	200,10	••••	957.44	2.206		29.00	41.40	231.61	1.44*
196	00.7V	*7.20	960.75	2.272	**	27.60	34.8((33.9)	2.051
347	401.60	\$7.50	769,29	2.246	27	29.90	\$4.60	235.11	4.04
348	202.90	42.00	949,41	2.001		30.10	37,90	235, **	5.700
350	294,54	38.20	775.73	8.100	90	\$1.00	24,40	239.39	1.465
121	285.20	42,70	982.75	2.175	21	31.50	27.00	241.43	1,927
433	201.10	38.50	143.9.	4.104	73	33.44	<3.80	248.71	1.477
128	207,8¥	37.80	908.70	2.093	24	\$3.00	45.00	250.26	1.896
156	204.24	-1.20	*** . 21	8.147	36	34.40	50.00	252.59	1.975
357	209.74	42.50	996.18 987 14	2.164	>7	34.54	29.00	252.45	1.959
					30	35.00	46.20	254.43	1.9%6
					•0	36.44	\$2.45	254.01	
101 0			.50 65.70		•1	36.40	38.00	240.57	2.006
					• 5	37.74	30.50	265.42	1.963
10. 6	1. B. C.	37,00 284	. 70 1.0"	.234		38.10	35.50	266.97	2,062
						37.04	42.00	279.47	2.006
					• ?		95.60	274.34	2,190
					•1	41.40	43.80	275.41	2,917
					10	41.50	48.90	285.14	2.030
					11	42.20	33.30	242.90	2 0 5 0
					11	**.**	\$7.70	287 45	2.097
						\$3.80	17.60	269.12	2,005
					1.	44,44	49.50	293.0(1.967
						43.54	34,00	295.72	2.117
					/* /*	47.00	38.60	296.81	2.111
						47.04	30,30	\$11.55	1.979
					01	4/.6V	15.40	374.65	2.060
					• 2	40.34	33.00 51.AD	306.60	2.022
					••	.8.74	\$3.40	305.15	2.024
					•5	44.00	34.30	309,32	2.077
					67	49.bu	34.30	310,10 312,41	2.010
					••	50.10	\$7.00	313.54	2.085
					49 10	50.44	37.90	314.76	2.098
					71	51.64	\$3,00	320.20	2.022
					45	52.50	36.42	1.1.1.	2.976
					9 J 74	53.00	53,30 58,00	324.46	2.027
					*5	59.10	46.00	329.15	1.911
					76	54.84	32.00	\$\$1.85	2,003
					78	55.40	46.30	334.18	1.727
					77	56.00	27.85	316.52	1.472
					100	36.46	25,50	356.07	1.903

		PEACE BONE	HULL D	AB-24			
J	NISITICE	0141112	10	COMPUTER	COMPUTED	ر	01511
	C-C+LH* 1	0045 i T T		06914	0245111		(/EP74
	(014)	(014)		171)	(6A/CC)		101
343	58.44	90,8¢		343.84	2.101	,	١.
145	99. DL	34.50		344,17	2.014	2	. د
103	57.54	57.80		290,11	2.066	5	٤.
* 11 %	60.0L	25,BA		\$52.06	2.065	•	۰.
142	£C.20	37.51		352.83	2.093	,	
I U P	&u. NU	36.00		355,14	2.070	•	ે.
103	61,00	34.00		355.94	2.038	2	<u>.</u>
108	61.30	29,00		357.11	8.022		
	67.00	37.00		339.93	2.085		
		33.60		101.38	2.037		
		37,40		363.71	2.092		
		37,00		366.04	2.137		;•
	45.66	31.20			2.007		
		*1.20			*,192 * 180		
	10.00	37.8		436 87	1 991	14	
	4- 36			3/3.3/	2 0 2 -		
	h	33.50		378 88	2.030		
	61			341.05	2.396		10
1.40	1.1.20			No. 97	2.093	- N	
44.3	50.11			363.33	2.334	41	
				387.62	2.390		
2	A * . *	31.50		308.20	2.093		12
1.41	**.*3	38.41		390.32	3.116		11.
	70.20	58,10		392.07	8.112	15	13.
606	7	20,00		398.24	2.20)		12.
147	18.24	46.00		345.57	1.411	٤)	15.
* < #	73.70	34,00		\$97,93	8.03#	**	15.
144	74.54	33.30		400.42	8.030	69	15.
	73.50	+7.00		483.73	5.082	30	34.
1 2 1	73.90	34,40		466,06	8.051	31	17.
125	14.20	57.70		907.23	2.047	*5	17.
1 2 2	74.80	37,80		489.76	2.094	3 5	18.
1.34	75.60	54,40		411.89	2,044	34	1.
132	75.64	\$6,00		\$12.67	2.070	35	17.
120	/8.30	37,00		*25, 54	8.085		20.
137	76.70	39.00		*16.94	2.117	• 1	20.
130	77.50	20,30		120.05	2.104		24
	74 70	37.60		**1. **	4.14/	37	21.
		AN			2 0 7 3		
1.92	79.84				2 915		
143	84.86	44.40			2.038		
1	93.68				7 180		23
1 * 5		39.40		\$55.20	2.054		
1.46	\$2.00	44.08			2.022	**	25
1.47	62.54	45.00		834 70	2.055	<u>ŝ</u> ,	
140	\$3.00			441.42	2.203		24
1 4 4	\$4.80	19.00		**3.30	2.317	••	26
120	29.5V	41.00		\$ 47.24	e.144	50	26
171	60.10	\$3,00		449.57	8.027	21	21
125	85.7V	29.00		951.91	1.959	52	26
1.75	AL, DU	24,00		\$53.07	1.080	23	28
124	P6.50	29,00		925.01	1.75*	34	28
1.75		5#.Qu		156.14	2.101	35	24.
						56	27.
						57	54
_	.					58	50,
1.1.4.	- 23		1.20	63,7)		24	30.
		30.00				•0	30.
		a0,00	e ; ;	1.00	********	•1	

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ل	DISITIZED	016111260	COMPUTED	CHEWITE C
	UEPTH, 2	DENSIIT. T	DEPTH	DLASITI
	IULVI	(01v)	(FT)	(64/00)
,	1.90	42.00	169.37	2.002
2	3.80	46.60	175,70	2.074
5	5.6U	\$5,20	376.06	2.052
:		36,73	179.21	2.075
	3,00	19.30	180.74	4.114
,	5.40	W0.00	183.15	4.127
•	3.00	38.50	103.94	2.103
	3,60	36.20	184.72	2,067
		36.00	287,00	2.055
12	7.60	38.60	189.95	8.185
13	7.50	36,20	190.63	2.099
	/	35.00	192.60	2.049
15	8 50	36.50	194,57	2.077
	9.00	39.00	193.35	2.111
18	9.40	43,40	198.90	2,180
49	10,00	\$5.00	201.26	2.205
24	10.60	•3.•2	\$03.65	2.108
	14.20	35.00	206.38	2.017
e 5	30,70	\$1.40	211.84	1.992
	13.30	34.00	214,25	2.053
13	13,60	34.20	\$15.43	2.136
.,	14.20	10.00	219.72 217 An	2.044
	14.89	44.30	228.14	2.038
41	35.30		222.13	2.041
30	17 40	36.30	224,88	2.069
32	17.40	28.70	220.02	1.450
3 3	10.00	52.60	232.76	2.011
34	18,44	\$3,50	235,12	2.025
35	17.50	40.30	237.07	1.944
	20.30	21,00 51.00	240,43	1.923
38	26.50	32.00	292.60	2.002
39	27.00	30,40	244.57	1.976
	21.50	29,00	2=6.54	1.955
	22.50	29.80	248,90	1.904
* 3	23,00	30, 30	252.44	1.975
	24.70	\$5.00	255.20	2.049
	24.20	35,20	257.17	2.032
47	25.80	\$7.20	260.31	2.083
÷#	26.00	35.60	269.25	2.058
• •	26,40	\$5,20	265,83	2,052
50	26.80	35.80	267.40	2.061
32	26.00	36.11	268,78	2.044
> 3	20.50	1 00	273.31	2.044
24	20,70	93.10	274.88	5.014
35	28.80	45,10	275.67	2.025
57	24.76	\$4.30	278.82	2.036
58	30,00	35,45	200,00	2.055
37	30.50		281.97	5.034
	30,80	33,80	243.15	2.017
•¥	82.84	49.30	249.06	1.962
• 3	34.70	28,30	290.63	1.944
	53.40	42.50	293,39	2.009
66	34.19	44.75	284.14	2.01*
÷,	34,70	48.00	298.50	2.017
• 4	54,9U	33.20	299,29	2.020
10	53.30 15 41	#3.20 #5.44	340.47	2.020
4	16.10	44,51	304.03	2.041
12	36.60	\$4.90	305.90	2.047
/3	57.00	42.00	307,54	2.014
	57,60	34.00	309,92	2.033
15	30.50	32.20	511.50	2.005
17	39.00	28,70	\$15.+3	1.4**
(*	39.40	\$2.00	317,01	5.00.
	50.10	40 50	310.76	1. 757
•1	40.40	33,90	370 94	2.011
• 2	40.90	\$1.30	322.91	1.991
* 3	\$2.0v	35.00	327.24	2.049
	42.60	40.00	329,61	2.127
86	43.80	25.30	334.34	2.055
#7	44,QU	52.00	\$35.12	2.002
		\$2.60	359.06	2.011
	43.60	83.40	341.42	2.055
¥1	46.90	40,70 40,50	544.57	2.035
45	47.20	\$1.30	347.72	1.991
73		51.30	350.87	1.991
74	48.50	33.50	352.05	2.025
*6	50.20	\$2.00	376.38 554 44	2.036
	50,70	29.60	\$61.50	1.964
78	51.30	29.70	363. *6	1.966
100	32.56	32.50	365.43	2.004
			200,20	*****

THE STATE STREET STREET

PEACE BUNEHOLE OFT-11

		PEACE BOREH	OLL 011-11	
J	n161714ED	21+11124	D COMPUTER	COMPUTED
	DC#141 X	DE 481774	1 DEP1H	DENSITY
	10341	10141	\$ # T }	16#/CC/
101	55.00	**.00	870.95	2.111
145	53.80	57,80	\$72,13	2,092
163	33.74	39,80	\$75,31	8.111
380	54,00	30,50	374,49	2,183
183	34.90	35,90	375,28	2.063
146	54.40	24,78	\$76.06	2.044
187	54,90	\$6.38	378.03	2,064
1.48	58.50	36,10	380,3*	2,366
109	52,94	37,20	391.47	2,065
110	36,30	30,84	393,54	2,096
***	36.64	33,90	384,72	2.063
115	56.90	47.98	885,91	2.094
2 4 3	57.94	\$6,80	387,48	2,096
114	57.60	45.30	588,66	2,053
115	50,50	33,20	391.42	5,050
216	58.99	35.40	393.78	2,055
337	59,64	39,80	396.54	2.111
110	40,0U	38,70	398.11	2.107
23.9	P8'A0	37.00	399.69	2,080
146	40.90	23,00	N01.65	2.044
141	61,74	36.00	414,80	2.064
142	62,00	37,20	\$95.98	2.683
143	62.30	34,50	\$07.17	4.072
144	63.00	35.00	409,92	2.002
145	\$3.40	35,80	*33.50	8.006
	84.00	33.80	*13. 8 6	8.058
241	64,3U	36,80	*15.8*	2.077
148	F# * 0A	33.00	N17,81	2.049
449	\$5,40	46,68	419.87	8.074
126	63.70	45,28	\$20.95	2,052
191	66.JU	36,60	*72.91	2.074
. 22	60,6¥	36.40	424.04	2.071
199	66,80	15,30	*7*,#*	2.055
10. 8.	s:	. 0 4	1.50 63.An	
	R. C:	30.00 2	1.0E	. 254

	•{	*:[80#["0[1	UK 1-1 '	
	015171460	0101112E0	COMPLETE	CORPUTES
•	UE#14. 1	SENSI11	Of Pite	CENSITY
	16151	16141	(F 1)	(66/(()
		41.50		2 1 50
;	4.24	40.60	1 92 . 1 1	2.156
	3.00	+1.10	185.26	2.144
5	7.64	\$7.20	203 47	1.926
•	. 79	40.50	267,71	1.978
2	9.50	24.70	210.86	1.807
	10.64	40 60	213.01	1 . 74
10	11.00	32.30	214.76	2.006
11	31.90	10,20	220.31	1,975
12	12.30	5= . 30	221,88	2.030
13	14.80	33.60	225,85	2.058
14	13.44	34.10	226.22	2.034
13	14.00	37.00	228.56	2,080
	13.00	37.00	232.77	2.000
	14.39	36.20	234.42	2.967
19	17.50	34.01	242.36	2.033
€ C	1'.00	1.70	243.54	1.497
¥1	19.00	57.60	248.27	2.009
42	20.20	+C.20	253.00	2.130
23	31 60	34.00	275.77	2.311
	22.54	\$7.90	26.2 05	2.084
46	25.80	47.20	267.17	1.926
e 7	24.6P	33,20	270.32	5.050
28	52.0V	\$1.00	271.90	1.986
	23.50	33.00	273.87	2.017
36	20,00	17.50	275.84	1.774
	27.54	33 30	241 71	2 036
	28.54		284.90	1.935
	29.60	26.40	240.02	1.919
*5	30.00	29,00	291.59	1.935
36	30.00	40.20	293.17	1,973
37	31.20	27,80	296.32	1,936
38	32.00	20.00	243.47	1.737
10	34.00	32.80	302 62	2.802
÷.	83.60	44,30	303.47	2.030
	33.54	42.40	305, 57	2.014
• 3	33,90	33.20	306.95	2.020
	54.50	32.50	309.31	2,004
	35.04	33.00	213.20	2.047
	34.54	35.40	317.19	2.455
4.	37.20	86,90	\$19.95	8,078
49	57.94	a5,80	392.20	2,061
		41.20	\$24.47	4.42
	17.80	43.90	328.00	2.031
	40.30	45.80	331.37	2.030
54	NO.70	41.30	\$33.73	1.991
35		41.00	\$34.57	1.986
26		29.60	335.30	1.964
28	92.99	30.30	330,00	1.970
51	** 20	49.00	347.51	1.933
	43.09	\$2.20	350.66	2.005
•1	43.64	\$1.60	393,03	1.995
• 2	·•.20	27.80	355,39	1.936
		29,00	357.36	1.955
	88.DU	35.00	360,31	2.047
	\$8.19	37.44	362.47	2.084
67	49.19	87.80	366.81	2.080
68	47,90	\$2.20	364 94	2.845
	34.34	53.00	371.54	2.017
	54.90	35.20	5*3.90	2 95e
12	52.10	39.20	378 43	4.037
15	54.54	55.60	380.20	2.050
14	33.00	\$5.60	382.17	2.054
(\$	53.10	45.10	382.56	2.050
1	33.20	\$6.00	382.96	2.064
	54.00	36.00	386,11	2.064
	54.19	26.39	386.90	2.072
	55.20	32.90	190 41	2.014
÷1	55.50	31.70	392.02	1.997
*2	55.80	43,40	\$93.20	2.017
	56.20	34.50	394.38	2.030
	30.50	34,80	395.95	2.045
	37.00	24.30	347.92	2.030
	56,68		277,89	4.047
	59.60	49.10	807.37	1.995
	60.00	43.30	409.75	1.900
¥0	60,90	24,90	113.28	1.890
	\$1.50	\$1.20	415.41	1,989
72	62,90	41.20	517.61	1,909
73	43.44	67.70 41	721.75	1.466
*5	64.10	\$1.70	124.70	2.020
96	64.50	81.00	427.44	1.986
	65,00	41.00	\$29.43	1.904
28	\$5.20		\$30.22	1.903
1	67,00	43,80		2.927
100		37,40	• 37, 34	1.014

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	PEACE RUREHOLE DET-13				PEACE BOREHOLE OKT-13					
J	018371420 Dipim. 1 (Ulv)	JI6171260 DENSITY - T (DIV)	CDMPUTEN DEPTH (Ft)	COMPUTED DENSITY (GR/CC1	ن	DIBITIZEL DEPTH. X IDIVI	01+111860 GENSITT+ 7 40141	COMPUTER DEPTH (FT)	[[##UTEE DE#\$171 [64/60]	
141	66.80	33,30	*36,52	2,022	281	132.40	45.30	694.87	2,850	
1.45	67.64	89.30	139.47	1.959	242	133.00	35.10	497.23	2.050	
782	68.10	30,50	**5.18	2.000	243	134.44	86.70	762.74	2.675	
105	67.40	45,20	446.76	2.052	242	135.00	\$6.70	705.11	2.075	
346	70,30 70 BU	36,00	458,38	2,864	70) 706	193,30	35,80 35,80	795,89 787,94	2.047	
100	71.34	38,00	*54.2*	2.096	204	133.90	37.70	708.45	2.091	
304	74.30	39,80	+57,39	2,124	404	136.70	37,20	711.80	2.043	
110	74.80	48,00	44.87	1,970	211	137.00		710.15	2.149	
112	75.00	34,00	46.81	2,033	475	138,30	39.30	716,10	2.116	
115	75,20	36,20	***.**	2,030	213	130.60	34,30	720.07	2.002	
145	76.24	44,20	475.54	2.856	415	142.10	42.00	733.07	2.150	
114	76.80	31,90 Am AC	473.90	3.045	216	142.00	41.80	755.82	2.155	
110	77,60	34,80	479.05	2.045	218	343.80	40.00	739.76	2.127	
119	78.20	58,00	481.41	2.096	219	145.50	*1.40	142.52	2.149	
141	79,50	32.00	486.53	2.002	221	146.60	35.70	745.45	2.044	
342	84.00	50,70	468.50	1,981	242	1 . 6 . 50	45,80	750.41	2.961	
223	80.30 80.80	32,00	488,90 991-63	2,022	223	247.00	38.30	752,87	2.100	
145	51,14	32,80	.92.83	4.01*	245	148.00	41.50	756.30	2.150	
446	81.50	35.90	495.62	2.063	206	348.70	*1.6 ^R	759.06	2.152	
140	84.90	44,68		2.047	241	149.90		768.74	2.108	
244		42,80	\$89,32	2.082	***	190.40		765.76	2.111	
131	83,60	32,00	502.68	2,002	230	191.20	36.00	768.91	2,108	
1.52	54,20	\$9.00	505.04	2.111	242	151.80	\$5.76	771.27	2.060	
125	85.00	48,30 34,10	510.17	2,034	433	153,50	34,50	777.96	2.041	
125	86.10	35.20	512.53	2,052	445	15: 34	41.20	785,85	2.146	
136	85,30	39.00	313,31	2,111	246	156.90	35.40	791.35	2.033	
1.00		40.70	517.65	2.107	444	197.14	34.20	792.14	2.047	
149		85,50 81.60	378.01	2,096	239	190.10	40.90	746.08	2.141	
141	90.64	\$0.30	527.87	2.135	(1)	137.40	48.60	801.99	2.185	
117	90,80	92,10	531.04	2,160	<u></u>	160.50	48.50	804.74	2.183	
144	91.60	\$3,50	534,19	2.182	444	161.60	43.10	010.65	2.014	
145	93.00	35.00	539.70	2.049	<u></u>	162.70	87.00	814,20	2.080	
147	45.10	57.60	347.97	2,089	25	169.16	91.10	419.71	2.134	
1.4	95. NU	38.00	550,73	2.096	474	165.00	48.70	423.25	2.107	
120	91.20	35.30	336.24	2.033	230	160.10	33,70	827.54	2.028	
151	90.00	34,30	559, 59	2,034	421	160.50	\$2.00	829.16	2.007	
129	97.84	36,30	543,35	2.067	(3)	164.14	35,50 \$8,57	835, 16	2.100	
134	99.20	\$7,76	564,12	8.071	236	168.80	38.20	\$38,22	2.099	
175	100.54	45,00	569.84	2.849	233 26	167.20	38.80	840,74	2.100	
197	101.10	37.40	571.60	8.086	437	1 10.80		642.95	2.104	
724	162.90	34,80	578.69	2.033	429	170.90	37.20	847.47	2.071	
1+0	105.94		579.87	8.852	500	171.60	44.50	849.25	2.103	
101	186.24	AN . NO	391.68	2,039	201	172.00	37.00	880.02	2.127	
185	106.00	36,40	594,85	2.071	463	173.20	44.30	435,55	2.100	
100	107.84	Ab. 70	596.02	2.875	444	179.30	\$8,00 (3.50	835,94	2.096	
1	200.94		399.96	8.887	254	175.50	\$0.40	864.61	2.184	
167	100.30 107.20	84,85 84.85	688,74 683.5P	2.045	267	176.20	\$3,80	867.36	2.817 1.984	
169	109.50	30,40	604.64	2,102	264	177.40	\$7.20	872.09	2.083	
1/0	110.8V 110.30	38,80 49.40	606.65	2.108	2/0	177.80	35.60	873.66	2,041	
172	111.20	\$3,40	411,30	8.024	472	178.70	43.70	877.21	2.028	
1/3	112.30	*1.70	635,71	2,154	213	179,30	35.00	879,57	2.049	
1/5	115.80	41.70	621.62	2.154	2/5	100.00	34.00 45.40	889,36 885.4A	2.212	
115	1150	*1.70	624.37	2.154	116	184.0V	\$5.56	090.20	2.054	
111	119.00	34,30	676.34 638.24	2.096	d/1 	183.JU 183.RU	37.8C 34 36	894,14	2.092	
119	117.10	38,70	634.61	2.107	2/9	144.50	34,80	900,05	2.045	
109	117.50	41,30	636.19	8.143	4*0	185.80	35.50	905.17	2.036	
10.	117.20		642.88	2.111	595	196.70	36,20	908.71	2.064	
103	119.00	39,90	6+5,25	2.125	e# 5	187.00	35.5C	909.90	2.025	
104	750.00	40.60	649.18	2,193	284	107,50	\$3,50	911.86	5.052	
146	121.10	\$6.70	\$ 50.34	2.075						
167	124.10	41,00	852.73 655.84	2,147	•	9. T		1.4		
449	125.40	\$2,90	459.42	2,172	101	- 3.		1.20 03.00		
190	129,20	43,10	462.57 514 0m	2.176	201	A+ B+ C:	29.54 21	1.00	. 23397	
172	126.00	\$5,00	667.66	8.030						
193	126.30	32,70	670.8×	2.013						
175	128.30	12,80	678,72	2,171						
176	129.10	38,20	\$81.87	2.099						
7.48	130.20	35,00	688.96	2,049						
199	131.40	36,80	6901	2.077						
200	12.10	3,3,10	693.69	2.019						

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		EACE BOREHOLL	0P2-18				CACE WOREHOLE	0#2-18	
J	016111460	0101717E0	COMPUTE	CURPUTLO	L	ni6111657	016111260	(Deputio	C
	(CPTN) 1	10451776 P	14 11 41	14411	•		35.441.11 1	DI PT"	
			,	10-2007		(614)	1014.	((64/22)
1	+6.84	9,00	387.01	2.70-	. 61	1.6.4 .6.9	17.60		7
2	8'.6U	7.00	393.83	1.964	103	105.44	42.00	611.94	2.187
	. P . OU	7.48	397,37	1.999	1 <i>43</i> 106	105.10	20.80	614.70	2.170
	68.6U 88.60	4.10	399.34	1.999	105	104.00	80.40	414,24	8.16*
;			452.00	1.915	106	104.86	19.60	621.34	2.131
:	47,9J	6.20	105.27	1.940	140	105.00	19.50	\$73.74	2.046
10	53.00	9,30	909,58	1.787	104	106.70	16.00	£26.87	2.094
11	51.59	7.30	411.55	1.789	111	107.20	10.20	\$30,84	2.129
- 23	54.30	15.20	18.63	2.002	112	108.00	23.30	633,49	2.204
14	55.60	14,90	*19.81	2-971	114	189.00	23.35	\$37,92	2.20*
34	55.50	16,78	26.91	2.105	112	109.60	22.00	640.29 643 24	2.149
	55.8V 56.8V	15,00	428.47	2.079	117	111.00	17.00	6-5.80	2.110
1.	56.50	11.70	. 50	2.027	119	111.40	17,90	647.37	2.124
40 61	96,80 58,30	15,00	*32,*1	2.04*	140	1120	17.70	6.5.40	2.121
	14. DU	20.00	441.07	8.157	745	115.00	16.20	657.97 661 55	2.124
43	59.20	17.60	11.44	2.119	143	115.60	16.20	663,91	
45	6U.00	13.50	***.1*	2.052	145	117,10	15.20	665,C*	2.082 2.080
47	62.8V	17.20	*50.92	2-113	146	110.20	40.10	474,14	1.154
	68.50	1		8.15-	14	114.50	15.46	678 47 680 84	2.044
30	63,94	20,60	**5.6*	8-167	144	150.00	1.00	681,23	2.041
21	64.30	11,30	*61.**	2.020	191	121.10	19,20	685.54	2.07
33	\$3.24	13,20	462.73 868 WB	2.050	1.00	124.14	18.00	689.30	8.126
**	\$3,40	15,00	\$66.27	2.074	194	123,10	23.30	643.44	2.204
35	66.84	12.60	***.2*	2.047	1 5 5	125.70	e*.20	\$93.80	2.223
37	67.10	13,20	\$72.96	2.050	11	129.50	29.20	678,10	2.146
39	68.09	20,00	474.51	2.114	1.00	125,70	14,90	703.67	2.140
•0	68.68	23,60	\$78.87	2.21+	1.00	. 26.70	19.50	703.64	2,171
1	67.50	22,70	479,26 101,67	5.500	191	127.20	19.50	709.58	2.1
• 3	14.04	17.20	101.38	2.113	1+3	154.00	17,10	712,33	2.107
* *	70.40	17.40	467.42	2.140	1.44	129.64	e0.00	719,03	7,157
**	71.20	10.90	449.10	8.140	1*5	1 30 60	19.60	722.94	8.191
	71.49	14,28	190.68	2.129	3+7	151.20	28,70	725.35	2.038
49	72,14	17.90	*92.65	2.174	144	154.90	16,10	732.02	2.127
23	12.85	14.20	173.10	2.060	121	155.00	15,00	733.20	2.091
72 73	73.00	13,30	*76.1*	2.083	125	154.10	14.20	736.74	2.097
-	13.64	14.20	\$98.55	2.046	174	153.20	20,00	799.84	2.157
37	75.00	10,00	501.31 504.07	4.000	122	1 55. 34	19.10	742.47	8.343
57	75.54	14,50	506.03	2.102	136	136.20	10,70	745,01	2.105
24	76.70	18,60	508,74	2,148	178	136.90	16.70	747,77	4.105
	77,10 77,10	14.40	\$12,35	8.138	100	138.00	15,00	752,10	2.079
	78.20	18.90	514.66	8.140	141	158.50	16.70	754.07	2.105
	79,50	14,30	521.78	2.034	1.5	139.40	18.90	757,61	2.140
• 9	#2.30	49.00	\$20,06	8.134	164	189,70	17.30	758,74	2.118
	88.28	18,48	528.67	8.152	144	140.60	14.10	762.33	2.96-
	04,00	18.00	\$15.56	8.126	167	141.30 141.40	14.10	765,09	2.964
10	44,9V	15,74	541.47	2,990 2,891	164	141.90	15.20	767.45	2.082
2		18.10	545.81	8.833	170	142.0V 142.70	14,69	767,84 770.40	2.072
15	\$\$,\$0 \$7,\$0	12,40	546.59	2,034	172	1+3.00	15.50	771.78	2.086
1	\$7.00	16.60	\$51.31	8.184	1/3	143.54	13.70	773.36	2.090
16		14,80	752.10 353.47	2.105 2.135	175	1	12.30	775.72	2.036
12	80.20	17.70	556.03	8.121	117	145,10	9.2D 11.20	778,87 780.64	1.967
1	A9.10	45,00	338.79 359.50	8,868 8,879	378	1=3.70	9.90	782.41	1.998
		15.30	348,76	2.083	140	146.4U	10.40	784,38 784,17	2.01× 2.013
	99.54	15,50	363,91	4.052	101	146,96	13,00	787.14	4.9.7
13	41.00	19.70	567.06	7.193	2#3	147.70	12.20	787,92 790,24	2.035
41	42.64	28.99	378, 79 373,36	8.252	784	1.48.10	14,00	791.84	2.063
	78.98	~2.00	374,34	2.109	144	150.00	10,00 17,50	796,19 799,3m	8.124 8.118
	44.74	15,10	376,90	2.347	107	190,80	14,70	802,49	2.07.
89	48.30	16.20	388.05	2.897	103	151,40	13.50	804,07 864 84	2.034
3	45.20	14.08	3#2.#1 383.3	2.212 7.342	170	192.00	16.20	807.21	8.09
*2	46.70	14.90	387.53	1.077	272	150.00	19,50	808,79 811.15	2.086
*5	47.nu	3.10	393.03	1.892	195	153,50	15.20	613.12	2.062
**	30.50 87.50	2.90	393.80	1.888	1 * 3	154.65	15,50	818.24	2.044
• 1	47.40	7.65	692.15	3.99%	146	158.20	16,90	419.41	8.108
**	108.90	11.70 11.60	4#3,28 606.81	8.627	170	196.20	12.20	443.75	8.885
Tra	747.50	14.00	607.21	1.894	500	197,30	10.88	826.51 878.98	2.000

		FLACE BOR		0P 2 - 1 P	
J		. •• • • •	•1.		
	DEPIN, R	00%911	1 · · ·	D[# 1#	OLNAITT
	(014)	(014	•	1993	168/001
	197.94			438.54	1.963
448	338,40	10.0	e	852.91	a,988
448	190,74	•.1	•	833.59	1,984
864	159.20	34.3		835,54	8.868
405	339.60	34,7	0	637.3%	8.874
146	364,80	16.3		030.71	2,099
297	361.04	24.6		842.65	2.072
448	162,84	14.2		846.39	4.864
489	162.64	13.0	0	618.95	8.947
210	263.54	18,8		852.59	2.060
<11	364.00	36.0		854, 16	2.89%
412	164.30	17.0	0	855,64	8.110
613	165,80	6.5	0	861.95	1.942
634	366,40	10.0	0	463.91	4.000
413	166.84	11.0	0	#65.48	2,916
415	167.20	10.0	5	867.04	2.000
447	168.00	13.2	n	870.23	2.050
418	160.20	13.2	e.	870,99	2.030
434	160.40	15.6	. C	871.78	2.091
160	369,35	15,7	C .	\$74,54	2.090
ed 1	169.60	14.0	(876.51	2.879
242	169.80	15.3	0	877,29	2.083
243	170.20	17.2	C	878,87	8.113
664	170.60	36.7	16	500.44	2.100
445	171.00	10.5	C	882.02	2.134
	5:	15.80	3.00	63.40	
(J. A.	B. C:	30.80	324.90	3.07	. 254

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1	Claitiffe	016111762	.t*·	-0=+('*t[
	(0191	101A1 DEMB1441 A	OF TI	(be/CC)
1	20.80 20.70	26.00	189,93	1,921
	21.20	46.00	194.75	1.915
	21.84	27.00	193,95	1.931
;	25.40	10.00	198,76	1.516
:	23,60	7.00	201.90	1.612
10	23.70	7.70	204.79	1.623
	24.50	17.80	206,40	1.637
11	24.80	16.00	209.21	£.755 1.742
45	25,10	15.10	210.41	1,701
17	25.40	14.80	211.62	1,768
4.9	26.40	2.30	214.03 215.44	1.659
4U 43	26.90	3.00	237.45	1.548
42	27.50	13.60	220.04	1.908
**	27,50	10.40	225.20	1.565
45	34,80 54,80	10.60	229.30	1.669
e7	51.0V	1.20	234.14	1.519
	31.60	5,50	236,53	1.561
80 31	51,90 52,90	5.10	237,73	1.581
37	32.50	. 70	240.14	1.511
34	55.40	13.20	243.76	1.711
35	55,70	18,80	244,94	1,800 1,690
37	54.6U	11.00	248.54	1,675
	33.60	36.00	252.60	1,755
	36,60	18,00	254,20 256,62	1.789
3	17.00 17.10	17.00	258,22	1,771
	57.70	<1.30	261.03	1.040
	34.29	19,00	263,04	1.817
	30,70	18.80 17,90	263.44	1.767
99 50	39.40	15.20	267.86	1.711
51	40.10	45.20	270.68	1.902
	41.80	22,50	272.6A 274,24	1.860
34 33	41.20	42.00 43.20	275,10	1,851
36	42.10	23,20	276.75	1.670
26	12.70	46.20	281.12	1.910
		23.20	284,34 285,34	1,870
• 2	44.30 44,70	₹3,80 ₹4,70	267.55	1,880
• 5		23.60	287.36	1.874
• 5		24.20	291.17	1,778
5.7	46.70	10.00	297,19	1,639
58	47.50	43.00	308.41	3.847
14	47.70	41.20	207'51	1.85
12	NB.20 NB.50	30,80 28,20	303.22	1.991
25	46,40 48,40	30.80	304.02	1.991
15	49.20	45.90	307.25	1.913
17	50.10	42.10 24.10	309,65	1.832
/ *	50.50	<0,30 <₹.60	312.46	4.824
40 11	51.54	29.70	\$16.96	1. 774
	34.20	41.00	314.29	2.010
*5	52.80	21.90	\$21.70 \$29.91	2.151
¥3	54.1"	40.00	324.92	1,978
87	54.90	29.90	330,13	1,977
	55.80	13.00	332.34	1.734
70	56.48 56.70	18.00	336.16	1.787
72	57.10	18.20	336.97	4.711
	50.00	20.60	341.79	1.829
74	47.00 57.00	42.80 43.60	546.61 348.21	1.864
97 V#	59,80 68,50	80.50	\$47.62	1.903
99 3 H A		29.00	355.44	1.899
	• • • • • •	46.6U	377.35	1,924

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J	DISITIKED JL+TM. L (DIV)	016171260 	20400755 36934 1573	CUMPUIEC DENSIIT (GR/CC)	L	D16111/(C D10144, 1 10141	(DIA) CENZILL L CENZILL L	сомынтен D26тн (ет)	100/0761 DENGITI 100/0761
101	67.30	26,60	379.95	1.92*	201	121.70		598,50	2.130
743	60.CU 60.5V	29,70	582,76	4.122	202	122.80	*2.60	602.92	2.227
1.0.4	69.00	31.00	386.70	2,007		123.10	84,50	604.12	2.050
144	78.30	>2.30	392.80	2.354	203	24.60	35.70	609.35	2.067
3U7 398	70,90 71,69	58,10 32,68	394,41	2.108	267	124.80	36.00	610,95	2.074
109	74.00		378.83	2.130	e U •	126.00	*6,60	\$15,77	2.047
771	75.24	49,80	483,65	2.122	230	176.80	47.00	618.94	2.090
112	75,50 75,80	48,60 44,60	484,86	2.308	412	127.20	37.0C	620,60	2.099
11.	74.38	42.70		4.022	415	170.11	36.20	624.21	2.077
115	75.10	47,00	\$14,90	2,090	213	129.20	52.5C	626,62	2.015
117	76.20	33,20 93.00	415.71 818.92	2,030	217	129.60	52 00	630,26	2,010
117	77.60	22.20	421,33	1.854	<i>4</i> 19	150.20	\$2.90	632.65	2.02
341	70.20 70,80	33.40	-26.15	2.031	440 441	131.00	21.90	638,67	4.033
142	77.50	44.00	428,96	2.202	667	132.79	>1.07	642,64 641 8/	1.004
		\$1.50	433,78	1.999	44 ⁴	133.99	32.90	4+7.51	2.025
145	61,30 82,90	31.80	436.70	1.693	225	134.4U 134.80	30,40	69,37	2.004
147	82.70		441.82	2.090	221	1.5.20	34.31	6.2.76	2.047
147	84.50	34.60	**5.83	2.052	247	130.10	34.30	656.35	2.047
4 2 1	84.UC 84.64	37.00	**7.0*	2,090	230 (31	136.90	32.90	639.76	5.050
1.12	65.0U	37.60	451,06	2,180	432	137.46	52.80	641.57	2.025
134	86.00	\$3.20	195.00	2.834	233	158.40	34.80	\$65.59	2.042
138	86.5¢ 86.8V	23,50 41.80	*57.89	2.007	235	159.40	33.00	667,60 669,63	2.050
1 37	87.34	86.08	468,30	2.87%	257	139.90	*6.50	671.62	2.082
1.39	80.10	46.70	463.51	2.005	439	1+1.14	\$4.30	676.54	2.050
141	67,94	25,00	467,13	1.699	241	142,20	\$7,90	680.86	2.304
142	87.98 98.0u	38.00	468,74 471,15	1,787 2,154	242	142.40	07.50 \$3.10	681,66 684,47	2.098
	99.50		*72.35	4.202		144.90	33.40	687.64	2.055
1.1	91.20	00.54	475.97	2.918	296	144.70	\$2.50	\$90.90	2.010
147	91.70 92.14	37.60 71.50	477,98 477,58	2.090	247	145.3U	\$2.30 \$4,00	691.70 693.31	2.015
19	92,70 88,24	36,30	401.94 883 AG	2.077	249	146.50	36.00	598.13	2.074
171	95.00	23.00	486.41	1,899	148	147.20		710.94	2.104
173	94,70	25.20		1,470	(3)	148.10	45.60	784.56	2.040
122	95.10	17.28	475,45	1.774	234	148.30	45.30 48.00	705.36	2,063
196	96.74 97.19	18.40	*98,86 *99,67	1.690	236	149.84	38,40	709.34	2.112
1.78	97.74	12.00	502.00	1.691	438	149.80	\$9.50	711 59	2.130
1=0		30.50	985,70	1.986	237 260	190.50	•0.00	714,20	2.136
1 . 5	97.30 100.60	22.00	909.31 513.73	1.851	/61 /62	150,90	99.70 A7 00	715.81	2.133
103	101.20	38.90	916.14	8,120	283	191.60	36.20	718.67	2.077
1+5	101.94	41.00	518.95	2.001	200	152.70	36,10	720.63	2.076
106	102.20	38,50 38,50	928,16 922,17	2.030	266	158.84 158.44	49,00 49,30	725.05	2.122
108	103.00	28.30	523,37	1,951	266	153.90	47.00	727.86	2.122
1 / 0	103.70	\$0.00	526.19	1.970	247	155.10	47.70	712.68	2 101
1/2	764'3n 764'0c	\$2.50	527,39 528,19	1,999	211	155.8v 156.70	39.70 36.80	735,49 739,11	2.133
1/5	104.50	31.10 31 TA	529.40	1,996	213	157.20	46.40	741.12	2.081
1	109.30	\$1.18	932,61	1.996	2/5	157.94	35,40	743,93	2.065
110	105.80	35.00	334,62	2.412	216	150,5U 158,7U	\$7.50 50.20	745.54	2.098 2.109
1/0	106.50	35.00 38.00	537.43	2.050	278	159.30	15.60	749.56	8.011
	107 9U		\$43.04	8.104	200	160.50	\$7.20	754.34	2.093
1.02	100.80	38.60	546.67	8.116	4+1	161.10	\$8.50 \$8,20	736.74 738.34	2.114
285	188,95 109,36	48.50 40.70	547.88 548.68	2.11*	283	161.80 162.80	37.50	739.60	2.098
1.05	110.00	37.20	531,50	2.873	203	162.70	\$6.50	763,22	2.002
	111.44	11 44	437.92	1.947	247 247	164.20	89,90 88,90	765.72	
144	113.20	43,70 \$0,00	760,74 564,35	1,978	200	169.9u 165.2u	13,30 34,40	772.05	2.021
1-0	113,80 219,40	42.90 48.88	566.76 567.17	2,023 1,757	278	163,70 *	44.50	775.27	2.050
1.45	115.00	** . 98	\$71.56	8.087	272	168,24	46.20	788.81	2.077
174	115.09	47.40	\$74.80	8.100	243 476	169.10	84.20 36.60	786,92	2.045
176	110.34	44,64	976,81 962,83	1,974	275 294	169,8V 170,30	47.50 46.90	791.74 798.74	2.076
744	110,30 117,19	29,48	584,84 586,86	3,969 3,911	497	170.70	40.30	793.35	2.111
140	190.40		573.28	2.111	299	174.80	48.40	199.77	8.114
					946	172.30	47.60	802.59	5.100

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	•	CACE BOREHOLE	###-81		PEALE BOREHOLE #49-01						
-	0++++++++ DEPTH ₁ 3 TU1++	01+11+2L0 3[45]11+ 7 40145	()480765 DEPTH (FT)	CUMPUTEC [LNS]TT (NM/CC)	ن	01617148 DEPTH, 1 16391		С ЧАРЈТЕ- ДЕРТН (Е Т)	CUMPUTEC PLN5111 (642:1)		
303	175.30	29.00	805 80	2							
542	113.01	48.00		2.106	405	230,90	20.00	1037.21	1.978		
384	174.10	34,40	809,01	2.049	403	231.90	30,50	1041.22	1.786		
385	175.00	35,00	919.24	2,038	404	232.50	46,00	1043.63	2.074		
847	176.00	34,20	A16.6.	2.045	446	233.24	\$3.50	1846.45	2.058		
508	176.50	\$2.00		2.010	407	233.6V 234 tu	\$2.90	1048.85	2.02*		
304	176.90	30,40 AD 10	626.26	1.983	449	234.70	\$2.50	1050.06	2.044		
811	178.00	**.10		1.948	410	255.00	47.50	1053.68	1.970		
413	178,70	99,90 40 90	427.49	2.042	*12	2344	32.CC	1057.64	8.414		
314	180.10	\$1,30	\$33.12	1.999	*13	237.40	\$5,10		2.060		
319	180.60	57.00	635.93	2.090	*15	230.00	42,50	1966.53	3,994		
317	101.50	\$6.00	\$37.9.	2.074	*16	239.20	\$2.40	1070.55	2.017		
318	101.00	31,80	***, *5	2.007	*18	240.20	•3.50	1073.76	2.196		
340	184.90	38.70	844.37	4.117	*19	240,70	45.7p	1076.50	2.229		
341	184.10	\$1.50	849.14	1.999	441	241.44	4,00	1074 44	4.202		
343	183,30	37.30	824.01	4.038	•22	241.90	49.40	1001.40	2.120		
5.00	185,70	36.80	\$15.62	2.074	424	242.50	36,70	1082.60	4.085		
3/1	186.30	\$2.90	856,82 856,03	2.045	*<5	243.00	\$7.50	1041.87	2.001		
367	167.30	37.40	812.0	2.120	147	244,30	36.0D 24 40	1107.02			
349	188.20	57.70	864.05	2.101	148	244.80	56.70	1093.05	2.044		
330	188.54	47.60	\$66.87	8.100	*27	203.10	34.00	1044.25	4.047		
335	189.50	39.20	A68.47	2.045	• • • •	243.64	36,10	1296.26	1.99%		
***	190.00	22.30	872.89	2.015	452	245.90	30,00	1097 57	1.978		
135	190.50	41.30	874,90	2.015	434	244.30	29,00	1098.2	1,966		
	191.20	30,90	877 7	1.995	***	241.00	49.30	1101.89	1.967		
337	192.60	38.00	380.93	2.106	437	248.20	\$4.30	1124.70	2.098		
339	193.20	33.70	885.75	2.037	869 769	248.50	33, ND	1107.91	2.033		
340	194.80	20,30 27,50	889,76	2.111	**0	249.10	40.10	1109,11	1.963		
342	195.10	39,00	A	4,042	4~1 443	544.00	40.10	1112.35	1.980		
	195.70	38.10 28.00	845.74	2.108	•••	250.70	31,00	1113.9%	2.039		
	197.00	**.00	901.01	4.202	***	231.20	32.30	1118.76	4.015		
	197.80	91.50	902.22	8.142	496	252.00	34,80	1120.37	2.022		
	198.00	\$9.30	905.03	2-127	447	252.50	35.00	1123.90	2.054		
350	199.20	49.50	907.84	8.130		294.00	41.50	1126.79	1.975		
\$71	199.50	43.50	911.06		450	255.60	41.00	1152 42	3		
323	340.40	43,30 88.80	912.66	2.063	*52	251.40	40.00	1134.03	2.001		
374	200.50	.2.00	115.48	4.170	* > \$	256.00	\$2.00	1138.04	2.010		
336	801.34	1,00	717.86	E.176	425	257,84	46.00	1140,06	2.014		
,,,	362.00		721.10	2.089	436	257,20	*6.30	11+2.86	4.042		
	202.94	49.10	929.71	2.120	• 7 •	258.DU	26.00	1145.67 2146.08	2.074		
3 m C 3 m 1	283,90 289,80	*3.10	784.73	8.187	***	258.50	35.00	11-4.09	2.054		
		\$7.40	981.95	2,096	**1	257.44	35.00	1150.00	2.090		
369	200.10 205.70	40.76	\$33,36	2.149	11	764.50	40,76 54,90	1152. *1	1,990		
202	206.10	40.90	\$37,57	2.146	***	261.00	42.40	1150,15	2.017		
347	206.60	47.80 30 80	948.38	8.127	166	261.90	84.20 87.10	1159.34	2.045		
3.8	207.44	11.20	742.80	2,157	467	262.50	90,54	1164.16	2.925		
3.0	208.1V 208.1V	44.6U 40.40	945.61	2.211		263,90	34.00 \$1.08	1166.17	4.042		
2/1	207.04	\$6.50	949.22	2.002	•/0	264.50	34.10	1172,19	2.044		
312	209,54	34.10	*\$1.25	2.044	-/1	265.40	34.00	1174,25	2.042		
514	\$11.44	28,90	735.63	2.151	*/3	263.00	34.50	1177.41	2.060		
315	e12.24	36,90	962.08	2.089		266.60	34,20	1179.82	2.045		
5.7	714.0U	27.30	767.30	2.001	\$76	267.00	34,40	1140.63	2.030		
3/8	214.70	34,30	972.12	2.047	4/7	267.40	15.20	1143.84	2.061		
300	210.94	38.00	975.34	2.074			36,71	1102.05	. 2.085		
291	216.50	38.70	979.35	2.117							
303	219,00	*7.50	986.18 989 80	2.258	7u. y.	S :	.04 1.90	£3			
	219.30		770.40	2,130	ett			•••.•0			
	220,40	Aw.00	995.02	2.106		a, c:	30.00 230.10	875.00	217.80		
387	224.10	44.60	1003.87	2.084							
309	275.75	#3.00 #2.50	1008.20	4.026							
348	224.50	40.00	1011.44	2.130							
372	223.00	40.60	1013.50	2.148							
373	220.04	+0.70	1017,52	1.990							
375	220,70	35,50	1020.33	2.034							
376	228.04	\$4.10	1029.55	2.04%							
148	278,74	37.00 40.30	1028.37	2.090							
344	231.0"		1233.44	<.018 <.018							
400	230.30	32.40	1035.60	2.017							

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APPENDIX 6-2

DENSITIES OF "CORAL" AND ITS SOLID COMPONENT AS CONTINUOUS FUNCTIONS OF DEPTH

Data on the composition of coral solids¹ place its density ρ_S in a narrow range of values (2.71 to 2.93 g/cc; see Tremba and Ristvet, 1986). Within that range, however, ρ_S varies erratically over the discrete set of borehole depths for which solid composition has been measured. Hence, in a given borehole, straight-line connections between measured (ρ_S ,z)-points embody virtually all the extant information on the continuous change of ρ_S with depth (or altitude) z. At that level of description, we have:

$$\rho_{S} = [(z-z_{m})\rho_{S}^{m+1} + (z_{m+1}-z)\rho_{S}^{m}]/(z_{m+1}-z_{m}) ; m=1,2,\ldots,M$$
 Eq. (13)

with (ρ_{S}^{m}, z_{m}) denoting point m of the set measured for a given lorehole, and with the points so ordered that depth decreases as m increases.

For the same borchole, let the z_j -points of Eq. (11) also be ordered so that depth decreases as j increases. Now, merge the two sets of z-values, and number different z's of the combined set in the order of decreasing depth (again), obtaining thereby the values z_k (k=1,2,...,K+1). The z-interval between z_k and z_{k+1} (k=1,2,...,K) must then lie entirely within one of the z-intervals on which ρ (the density of coral) has the linear depth-dependence ence specified by Eq. (11); it must also lie entirely within one of the z-intervals of linear variation of ρ_S (the density of the coral solid) specified by Eq. (13). However, a given z_k -value need not appear among the z_j 's of Eq. (11); if not, then, at $z=z_k$, the value of ρ (= ρ_k) is obtained from Eq. (11) for the z_j -interval in which $z=z_k$ falls. Likewise, for a z_k -value not among the z_m 's of Eq. (13), we find the z_m -interval in which z_k falls, and use Eq. (13) to compute ρ_S at z_k ($\rho_S=\rho_S^{\circ}$). Eqs. (11) and (13) can then be replaced by the following equivalent relations:

$$\rho = [(z-z_k) \rho_{k+1}^{-} + (z_{k+1}-z) \rho_{k}^{+}]/(z_{k+1}-z_k) ; k=1,2,...,K$$
Eq. (14)
$$\rho_{S} = [(z-z_k) \rho_{S}^{k+1} + (z_{k+1}-z) \rho_{S}]/(z_{k+1}-z_k)$$

In accord with Appendix 6-1, $\rho_k = \rho_k = \rho_k$ for the Y-Y profiles; for the BHG profiles, $\rho_k^+ = \rho_{k+1} = \rho_{k+1/2}$ where $\rho_{k+1/2} = \rho_{j+1/2}$ -- the value of j being set by the requirement that the interval from z_k to z_{k+1} lie on the interval of Table 6-7 from z_j to z_{j+1} .

For each borehole included in the BHG survey, Table 6-9 (located in this Appendix) presents the depths z_k (k=1,2,...,K+1) that mark the endpoints of

¹ See footnote 3 on page 6-1 for explanation of use of "coral" in this text.

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the depth-intervals on which both ρ and ρ_S are simple linear functions of z [Eq. (14)]. For each depth z_k , the density of either coral or its solid component is also shown (third and fourth columns of table). Where blank spaces appear between two listed values of coral density ρ (called "BHG DENSITY" in tbl. 6-9), those two ρ 's are identical; that same value applies everywhere between them. Also, where values of ρ_S at that blank spot is given by Eq. (13) with z equal to the depth listed (or to its negative, if z denotes height).

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MASS OF SOLID AS A CONTINUOUS FUNCTION OF DEPTH IN A BOREHOLE

Table 6-9 (in this Appendix 6-2) and Eqs. (13) allow ρ and ρ_S to be computed for any depth and borehole covered by the BHG survey. Hence, the mass m_s of coral solid can also be computed from Eq. (5). In fact, given piecewise linear dependence of ρ and ρ_S on z [Eq. (14)], m_s can be expressed in terms of z using only elementary functions. In particular, if z lies between z_k and z_{k+1} , then:

$$\mathbf{m}_{S} = \mathbf{m}_{S}^{k} + \int_{\mathbf{z}_{k}}^{\mathbf{z}} \rho_{S} \left(\frac{\rho - \rho_{L}}{\rho_{S} - \rho_{L}} \right) d\mathbf{h} \quad ; \quad k = 1, 2, ..., K \qquad \text{Eq. (15)}$$

where $m_s^1 = 0$ and

$$m_{S}^{k} = m_{S}^{k-1} + \int_{z_{k-1}}^{z_{k}} \rho_{S}\left(\frac{\rho - \rho_{L}}{\rho_{S} - \rho_{L}}\right) dh$$
; $k = 2, 3, ..., K+1$ Eq. (16)

Next, observing that $\rho_S/(\rho_S-\rho_L)=(\rho_S-\rho_L+\rho_L)/(\rho_S-\rho_L)$, we can write Eq. (15) as follows:

$$\mathbf{m}_{\mathbf{S}} = \mathbf{m}_{\mathbf{S}}^{\mathbf{K}} + \int_{\mathbf{z}_{\mathbf{K}}}^{\mathbf{z}} \left[(\rho - \rho_{\mathbf{L}}) + \rho_{\mathbf{L}} \left(\frac{\rho - \rho_{\mathbf{L}}}{\rho_{\mathbf{S}} - \rho_{\mathbf{L}}} \right) \right] dh \qquad \text{Eq. (17)}$$

Replacing ρ and ρ_S in Eq. (17) by their linear equivalents in terms of z [Eq. (14)], we obtain

$$m_{S} = m_{S}^{k} + \int_{z_{k}}^{z} \left\{ a_{k} + b_{k}(h-z_{k}) + \rho_{L} \left[\frac{a_{k}+b_{k}(h-z_{k})}{a_{k}^{\prime}+b_{k}^{\prime}(h-z_{k})} \right] \right\} dh$$
 Eq. (18)

where

$$a_{k}=\rho_{k}-\rho_{L}$$
, $b_{k}=(\rho_{k+1}-\rho_{k})/(z_{k+1}-z_{k})$; $a'_{k}=\rho_{S}-\rho_{L}$, $b'_{k}=(\rho^{k+1}-\rho_{S}^{k})/(z_{k+1}-z_{k})$ Eq. (19)

The integral of Eq. (17) is readily found as an explicit function of z (Pierce, 1929)

$$\mathbf{m}_{S} = \mathbf{m}_{S}^{k} + \mathbf{a}_{k}\mathbf{x} + \mathbf{b}_{k}\mathbf{x}^{2} + \rho_{L} \left[\frac{\mathbf{b}_{k}}{\mathbf{b}_{k}^{i}} + \frac{(\mathbf{a}_{k}\mathbf{b}_{k}^{i} - \mathbf{a}_{k}^{i}\mathbf{b}_{k})}{(\mathbf{b}_{k}^{i})^{2}} \cdot \ln\left(\frac{\mathbf{a}_{k}^{i} + \mathbf{b}_{k}^{i}\mathbf{x}}{\mathbf{a}_{k}^{i}}\right) \right] \quad Eq. (20)$$

where $x = z - z_k$.

From Eq. (16), it follows that $m_S = m_S^{k+1}$, when z is set equal to z_{k+1} in Eq. (20); with m_S^{k+1} known, a similar computation then gives m_S^{k+2} . etc. Thus, using the depths and densities of Table 6-9, the m_S^k -values (k=2,...,K+1) can be computed in sequence from Eq. (20). For each borehole of the BHG survey, the resulting values of m_S^2 , m_S^3 , m_S^{K+1} appear in the seventh column of Table 6-9, starting at the greatest depth logged in that hole. Listed in the sixth column is an approximation to $m^k - - m_S$, say -- obtained by setting ρ_S equal to its mid-value $\rho_S^k \equiv 1/2(\rho_S^{k+1} + \rho_S^k)$, the mean of ρ_S on the z-interval from z_k to z_{k+1} . Using that value for ρ_S in the integral of Eq. (15), but with ρ still related to z by Eq. (14), we can write

$$\mathbf{m}_{S} \mathcal{H} \underline{\mathbf{m}}_{S} = \underline{\mathbf{m}}_{S}^{k} + \frac{\underline{\rho}_{S}^{k}}{\underline{\rho}_{S}^{k} - \rho_{L}} \int_{\mathbf{k}}^{z} (\rho - \rho_{L}) d\mathbf{h} = \underline{\mathbf{m}}_{S}^{k} + (a_{k}x + b_{k}x^{2})\underline{\rho}_{S}^{k}/(\underline{\rho}_{S}^{k} - \rho_{L})$$
Eq. (21)

where $\underline{m}_{S}^{l} = 0$, and \underline{m}_{S}^{k+l} is the value obtained for \underline{m}_{S} by setting z equal to z_{k+1} in Eq. (21) (or $x = z_{k+1} - z_{k}$). Replacing the $\underline{\rho}_{S}^{k}$ (k=1,2,...,K) of Eq. (21), for all k, by the single value 2.821 g/cc (half the sum of aragonite and calcite densities) yields yet a coarser estimate of m_{S} , and the approximate values of m_{c}^{k} listed in the fifth column of Table 6-9.

Table 6-10 (also located in Appendix 6-2), identical in format and derivation to Table 6-9, differs from Table 6-9 only in that the density profiles on which it rests came from $\gamma-\gamma$ logging (not from BHG surveys).

DENSIFICATION: THICKNESS CHANGES AS CONTINUOUS FUNCTIONS OF DEPTH IN CRATER HOLES

Eq. (20) allows us to compute the solid mass per unit cross-section, m_S^a , from the greatest depth logged to the depth at $z=z_a$ in a given borehole. The horizon at z_a in the given borehole -- horizon "a" -- will generally have somewhat different depth in a second borehole¹. Let $z=z_a^*$ at that horizon in the latter hole, with m_S^{a*} denoting solid mass (per unit cross-section) in the second hole, from the greatest depth logged therein to horizon "a". For any

¹ As needed, the depths to a given horizon in different boreholes have been found herein by linear interpolation among the horizon-depths fixed by PEACE Program geologists (see Chapter 7 of the current Report, particularly tbls. 7-2 and 7-4).

depth above "a" in the first borehole, the solid mass between "a" and that depth is equal to $m_S - m_S^a$ ($\equiv \Delta m_S$), where m_S is computed to the depth in question from Eq. (20), using the a_k - and b_k -values for that first borehole. Likewise, in the second borehole, the solid mass from "a" to a given depth above "a" is equal to $m_S - m_S^{a*}$ ($\equiv \Delta m_S^*$), with m_S computed from Eq. (20) using a_k and b_k -values appropriate to that borehole. For each choice of z in the second hole -- $z=z^*$, say -- Δm_S^* can be computed simply by evaluating the right-hand member of Eq. (20) and subtracting m_S^{a*} from the result. However, in the first hole, finding the value of z at which the solid mass Δm_S above "a" is equal to Δm_S^* , plainly requires equating Δm_S to Δm_S^* . The relation determining z is therefore the following:

$$\Delta m_{S}^{*} + m_{S}^{a} = m_{S}^{k} + a_{k}x + b_{k}x^{2} + \rho_{L} \left[\frac{b_{k}x}{b_{k}^{*}} + \frac{(a_{k}b'-a'b)}{(b_{k}^{*})^{2}} - \ln\left(\frac{a'+b'x}{a_{k}^{*}}\right) \right] \quad Eq. (22)$$

where $x = z - z_k$.

Eq. (22), which is transcendental, can be solved for x (hence z) by numerical means. Exactly one value of x satisfies it because u_{S} [Eq. (20)] increases monotonically as depth decreases, and the mass Δm_S^* above horizon "a" in the second borehole is ≥ 0 . Solution of Eq. (22) for x is greatly expedited by foreknowledge of m_k (k=1,2,...,K): As k is increased from k=1, it reaches a level at which positive values of $\Delta m_S^* + m_S^a - m_S^k$ turn negative; the root of Eq. (21) must lie on the z_k -interval over which that change of sign occurs. With the root of Eq. (22) so bounded, it can be found easily by search or iteration. Repeating the process for a series of ever-shallower z's gives a set of (z^*, z) pairs for which the solid mass between horizon "a" $(z=z_a^*)$ and level z^* in the second hole is equal to that between "a" $(z=z_2)$ and level z in the first. The height of the column between z_a and z is $z-z_a$ (or its negative if z denotes depth); $z^*-z_a^*$ gives the corresponding height in the second hole (for equal mass above "a"). If the two holes are actually the same, but with pre- and post-shot density profiles representing the "second" and "first" holes, respectively, then $z-z_a$ and $z^*-z_a^*$ (or their negatives) give the postand pre-shot thicknesses of a column above "a" that contains the same mass of solid at both times. The change in that column's thickness due to shotinduced changes in density is just $(z^*-z_a^*)-(z-z_a)$ (or its negative).

Under the key assumption given previously on pages 6-7 and 6-8, the preshot profiles are found today in control holes -- whence, we do in fact compute (z^*, z) pairs from profiles in different holes. A detail of the calculation (noted previously on pages 6-8 through 6-10) lies in redefining horizon "a" at each successive geologic horizon met along the stepwise march in z^* (from depth toward the sea floor). With z^* referring to the control hole, z is then allowed to shift suddenly to its value, in the crater hole, at a newly encountered geologic horizon. Geologic horizons are thereby strictly retained as Lagrangian surfaces, regardless of departures from the ideal of simple subsidence, or of actual differences between pre-shot density profiles and control-hole profiles. In addition (see p.6-8 through 6-10), differences in column height are thereby computed from densities in materials that are (as nearly as possible) the same.¹

Curves of thickness-change vs. depth are plotted as a series of dots when z^* refers to a control hole (jumps in z can then occur, marking shifts to geologic crater-hole horizons). When z^* refers to a crater hole, curves of thickness-change vs. depth are drawn with dashes (jumps in z then mark shifts to geologic horizons in control holes). The mean of a dotted and dashed curve is also drawn, as a continuous line.

As functions of present crater-hole depth, the thickness changes computed from profiles of the BHG survey were presented in preceeding sections (p. 6-7 and 6-8 and 6-8 through 6-10) (see figs. 6-4 through 6-9). Corresponding curves, deduced from $\gamma-\gamma$ density profiles, appear on succeeding pages as Figures 6-27 through 6-53.

REFERENCES CITED

See pages 6-35 and 6-36 for references cited in this Appendix.

¹ Thickness-change curves were first computed with horizon "a" fixed near Contour D. Except for larger gaps between end-of-data and the sea floor, there are no appreciable differences between those curves and the ones presented in this report -- and no change at all in conclusions drawn from them (conclusions first reached, in fact, with horizon "a" fixed).

			BOACHOLE	U08-1'						5JRE 40L1	00#+11		
	PLACE	9H6	DE 15177	CUMUL#11+E		(CP. ****2+		PLACE		DE +\$177	LIMULATION	10.10 BASS	
-	0(* 1)	(54/10)	COMPONENT COMPONENT	AM6=7.871	800±68403	SHL TH	•	01511	DE 45179	DF SOLL'			LINEAN SPLINE
						FOR HW?				154/([FC# #40
۷	1890.00		2.0210				101	289.UL	2 036				31716
;	1140.81		2,8210				102	249.00	1		7813	27134	277.44
•	1134,74		5. 2144				10.	275.20		2.746	27926.	·*#51.	27051.
•	1120.30		8.7393				105	271.00	1.662	• •	28552	24471	28471
•	1102.80		2.7344				107	264.60	1.943	7.8341	28552	20477.	28471
é	1080.30		2.1242				100	293.40		2.8434	29258	24161	29161.
10	1065.30		2.7262				11.	257.00	1.947		1004 1	24963.	29763.
- 11	1012.00		2. 7303				114	235,10		2.8655	30192	Sn109.	50119.
12	1037.20		2,7384				114	812.00	1.447		31141	51047.	31062
1.	1013.40		2 11 10				113	207.20		2.4445			
3.	973.20		2,7242				114	186.30		4.4343			
- 12	990,90		2,0633				110	184.20		2,65#8			
17	962.70		2.7808				114	131.40		2.8735			
21	939,20 921,40		2,8945				121	148.20		2.034)			
24	912.20		2.0928				122	128.30		2.8365			
24	\$#7.ev	4.015	2,9183	45.	0.	n.	124	113.00		2.9275			
23	870,2L	2	2,9159	-	466.	966 .	120	86.20		2.4252			
21	657.00	2.018		1697.	1579.	1579.	121	66.90		2.9240			
24	852,40		2,9206	1016.	1744.	1784.	127	56.49		2.0210			
34	827.00	2.01#		3836.	2974,	2978,	130			5.0210			
	826.40	2.036	2.9292	3034. 3074.	2978. 1816.	2978.							
80 57	812,40 799.40		2.9743	3747.	3676.	\$676.							
\$5	789.00	2.036			.783.	4743.							
51	785.44	1.973	2.9082	4876. 3952.	6783. 4996.	4785.							
58	762.60		7.9246			5747.							
	747,00	1	•.•.•	6973.	4675	6644							
	787,80	8.015	2,91#3	6823.	K695.	6695. 71 81							
	725.10		2.8436	7725.	7779.	,,,,,							
4.8	718.40	8.018		**32.						BURLMOLE	044-41		
	F83'AA	2.018	8,844	9040. 9848.	8879.	8879.	•	PEALL	8H6 DINS111	0E45117	C SHUL AT J VI	SDE IN MASS	(BP/CH
	683,00 666.00	2.034		7658.	94 86 .	9686.		OFTI	1-4/611	LOWPONENT		RHORK#405	4P, INE
50	652.70		8.8768	11396.	11354.	13134.				154/6()			LCB WHG
54	645.40	1.993		11648.	11502.	11503.	0						
59	628,30 618.60		2.4588	12475.	1	12270,	•	• 52 . 71		2,8210			
33	.05.00	3.993		3846	19431.	139.11	- 1		1,905	5.9435	£.,	۶.	٤.
57	402.10	1.4.1	2.8997	13646. 13640.	13471.	19431.	;	421.90 813 80		2 7782	415.	117	117.
5 B	586,dv		2.8545	1	1-188.	1-1-1	•	.0	1.406		421.	121.	• • •
60	560.00	1.976	•••	15604.	14401.	19901.	Ú.	401.ev	1,422	2.7760	171.	429.	*/*.
	560,00	1.4/4	2. 72+2	15604	19401.	19401.	, 9	384.UV	1.972		1700.	1744	1744
**	433,60		1.7945		14625.	14625.	14	368.00	1.007	2.8307	2649.	7663.	2665
67	520.00	2.027		17059.	16884.	16884.	1.	354,00 354,00	1.849		3021.	3034.	3034.
	525.90 400.ev		8.7364 8.7889	17160.	16987.	16787	3.0	346.00		8.45*1	\$320	3350.	3330.
4.4	805.UL 806 0L	2.027		14163	14759.	10001.	i.	324.00	3.890				
7.	502.60	••	2.7848	18272.	14120	14120	10	380.40	1	8	4443.	4267. 4397.	4869.
14	491.40		2.7262	18743.	10137.	18149,	14	384.20	1.888	2.0476	5083.		
76	481.dv 477.dv	1.987	2.7242	19853.	14120.	19120.	21	294.00	1. ** 5		5505.	4306.	5566.
75	677.00	1.976		19444.	19317.	19517	2.5	270.70		2.8144	5457.	5637.	5639
11	456.20		2.7241	20393.	14633.	19653,	23	264,00	1.995			6834.	\$2.14.
7.	446.UU	1.976		19898.	20759.	28749	2.	260,54		2,9613	4775.	6982.	6982
		1.70	2.7181	21975.	28734.	20744.	20	294.90	1.913	2.950 0	7426.	7427.	7459
87	420.4U		2.6777	\$1751. \$2815.	21664.	21644	**	234,00	1.837		6101.	.171.	8101.
	617.UH	1.407		22166.	22082	22892,	31	289.00	1.037	4,3741	9073.	86F4. 4066.	8684. 5064.
83	411.76		2.72+2	22166.	22087. 22319.	22002.	32 38	209.00 208.70	1.845	2	9075.	.066.	9066.
86	400,70 588.56		2.7476	72885.	29812.	22412.	14	190.00		2,854;	9821.	4076. 4889.	98098. 9809.
	588.44	1.944	1.01.23	>3443.	23333. 23375.	23351.	32	184.00	1.945		100-7	10444	10044
87 90	588.UU 176.eu	1.451	2.7467	23443.	23375.	23375	57	176.00		2.8168	10400.	10347.	10347
91	363,00	1.451	•• -•	24 352	24491	24541,	37	139.00	1.920		11129.	11115.	1111*.
44	363.00 574,20	7.4.5	2.93 %	24352.	24441,	24441,	4U 41	146,00		2,8409	13724		1171
79	345.00	1.872	-	25282.	24210.	25210.	42	131.00		4.9946	17483.	12263.	17763.
-	344.48		2.9286	25816.	25210. 25243.	23210,	••	115.00 84,20		2,927# 2,927#			
	373.6V 317.00	1.901	2,4565	25970	25889	25867	43	84,19		5.8510			
99	317.40			26337	26373,	26373,		.00		2.8210			
100	277.00		5,1425	; MO1.	27319.	27819.							

TABLE 6-9. -- Mass of solid in vertical columns of unit cross-section, from BHG-survey data (continued on next 3 pages).

U	1808,00		2.8210				ť	1400.00		7.8215			
1	591.41		5.8510				1	018.51		2.9210			
	571.40		2, 8678	•		в.		828.50		2,9096			
τ.	572.54	1.747	2 2571	A12.	875.	375.	3			2,9191			
	551.94		7242	1468.	1274.	1278.	:	798.10		2,9042			
	551.44	1.929		1899.	1314.	1314.		796,00	1.430		4.4.7		
1	537.00		2,7242	1488.	1919.	2414.	7	768.84		2.92 17	1311.	1287	1247
	527.44		2,7241	2021.	2361.	2361.	•	768.00	1.956	•	1398.	1 1 3 3	1\$18.
	221.00	1.989		7798.			٧	768,00	2.041		1330.	1333.	1413.
17	911.84		7.7190	8019.	1047	1469.	1.4	759,10		2.4148	1771.	1758.	1734.
14	4 41.44	1.949		\$476.		+ Dnc .		744.16		2.4113	2247.	2214.	5514
3.0	491,90	2.009		3726.		NOn0.		738.86	2.011		2798.		2747.
19	+83.60		2.7242	4369,	4453.	***3.	1.	729.34		2 91 59	3199.		
13		2.809		****	• 963 .		12	708.00	8.011		4714.	4139.	4.19.
1.		8.040				\$215	1 *	708,00	2.010		4234.	4159.	4144.
			•••••	5848.		1944		700.10		2,8474	4590.	4404.	4584.
19	431.UL	2.01.		5646.		1964		575,00 175 JU	2.010		3941.		
2.0	442,80		2,7141	6437.	A361.	6362.	20	.70.00	1.013	2. 31.06	40.20		3344,
21	#35.18		2.7340	• 7 • 5 •	687 % .		21	637.30		2.8931	6516.	6.05	
54	#21.UV	2.014		7269.	7416.	(11 .	22	648,40	2.013		7041.	6945.	6945.
23	621.00	2,003		7567.	7414.	76.2	23	P## 00	2.030		7061.	6945.	6943.
5	501.94		8.4146	\$165.	371	\$321.	<u> </u>	629.44		2,7242	7956.	7849.	16.9.
26	371,00	2.003		8674.	8827.	8829,	23	618.00	2.030			8407.	8407.
21	347.00	1.9**		8674.	8429.	E#29.	57	578.60	2.040	2 8081	354A	A194	
50	367.70		2.997%	8827,	#980.	P980.	2.0	388.44	2.040		9943.		9873
27	367.40		2,0341		9909.	16203	29	584.90	1.951		9963.	9873,	9873.
	301.00	1.707		10440	10203.	10203	5 U	578,50		5.8343	10300.	10294.	10294.
12	131.44	1	2.9978	19585	18622.	10622.	31	558,00	1.951		11294.	11208.	11238.
3.0	352.40		2.0502	11324	11449.	11447.	34	558.00	1.944		11294,	11206.	11274.
34	331.00	3.998		11386.	11910.	11510.	39	538.90		2 71 90	17144	11613.	11013
35	331.00	1.455		11306.	11510.	11910.	35	528.00	1.948		12420	12562	12947
36	313.00		3.7844	12196.	18249.	17744.	36	528.00	3.979		12670.	12362.	12542.
37	343.34		4,0403	12476.	19700.	12799.	37	515.00		2.71≪0	1 \$202.	13154.	19158.
3.9	341.00	1.992		12475.	12799.	12799.	30		1.998		1#018.	13995.	13993.
48	288.04		8.6277	18+38.	13554.	13944.	37		2.030		1.010	14993.	13993.
93	\$12.44		8.8545	19731.	14052.	140 57.			2.05.	4. /1-1		16725.	1992
	\$71.44	1.948		12993.	10113.	14113.	44		2.046	2.4042	15999.	1** 93.	154=3.
	371.00	1.007		3 3 4 9 3 .	10113.					2.7241	16477.	16474.	16474.
	234.88	1.825		19994	14673.	19673.		433.00	2.046		17811.	17821.	17221.
46	234,48		8,0341	19923.	10742.	147+2.		633.00	2,011		17811.	17271.	17821.
47	285,60		8,8683	19921.	15635.	19655.			2.011			1	100-0
	224.00	1.926		12844	14959.	13959.			2.004		18627.	18698	18448.
47	358.00	1,971		33849.	15957.	17757.	47	379 50		2,8905	14792.	18511.	10011.
	211.90		2.8839	16211.	14412	16617.	34	380,54		2,9840	19687.	14693.	19693.
58	211.00	1.973	••••	16529.	14430.	10630.	51	300,00	2.004		19710.	14716.	19716.
58	211.00	1.026		16529.	14630.	36630.	24	300.00	1.922		1 7 10	19716.	19716.
59	200.44		8,8036	16916.	17012.	17012.	54	458.00	1.972		20556.	20646	20646.
32	140.40		2,7849	17319.	17414.	17414.	55	355.90		2.9040	28769.	20759	20758.
20	100.00	1.045			17304,	17585	56	\$38,70		2.8365	21531.	21513.	2151 1.
50	178.19		8.7070	17961.	17953.	17953.	57	328.00	1.972		22016.	21997.	21997.
59	165,90		2.6299	18436.	18525.	18523.		328,00	1.990		22416.	21997.	21947.
60	363.14		2,9178	18663.	18747.	18747.		311.44		2. / 4/3	22/63.	22746.	22746.
61	100.00	2.010		10715.	14798.	18798.	61	298,00	1.900		28345	23.527	23327.
		1.040		10/15.	10790.	19314	64	246.30		2.8233	28485.	23589.	235a*.
14	386.00	1.458	••••	19641	19726	19726.	6.5	275.00	1.900		2-890.	24 571.	24371.
62	1 51 . 74	•••	8.9096		•••••			275.00	1.906		24 840.	24375.	24\$71.
	131.07		8.8210					251,90		2.8/08	23281.	23234.	79254.
•/			5.0210					238.44	1.890		23847		23034.
								226.90		2.8798	26327	24290.	26290.
							67	216.50		2.0541	26766.	76725.	26725.
							74	208,80	1.894		27310.	27868.	27048.
							71	208.00	1.945		27316.	27868.	270KP .
							7.4	198.90		F. 8079	21398.	27326.	27556.
							7.	178.00	1.942		28124.	27066.	28375
							75	178.00	1.989		28428.	28378.	28378.
							76	171.20		2.8347	28726.	28674	28675
							**	168.00		8,9139	28866.	24813.	20013.
							76	163.40		2,9133	29867.	24810.	29018.
							8.P	184.99	1.939	2.9213	27/11.	74647.	27642.
								148.00	1.607		29741.	24672.	29672.
							84	180.00	1.007		30116.	30039.	30039.
							83	12. 30		8.9891			
								122.90		2.9145			
							85	118.00		8.9197			
								337. **		2,8210			

PEALL DEPT-DIFT-

*

CHMULATIVE SULID WASS (GP/CHARZ) LINGEF HMUEI-821 RH #CAMOS SPLINE FOR AND

DENSITT DF SULIC CDWPDNENF ISH/LCF

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9386406E 0H1-20

PEACE NEPTH LAFTI

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846 DENSITT 164/111

2545174 CUBULATIVE RULID WASS (6775+02) 17 5010 CUBPORENT PHUE2-021 RHOBERMOS SPLINE (5475) FUR RHO

TABLE 6-9 (page 2 of 3 pages). -- Mass of solid in vertical columns of unit cross-section, from BHG-survey data.

			8146-016	076-93						BORLHOLE	0#2-1+		
	PEACL	8+45	0[+\$171	CIMULATEV!	50-10 ##55	1.4/5=+421		PEACE	8-6	DENSITY	COMULETING	SU-10 -455	- 60 / 5 40 4 7
*	21431	DENGLEE ISN/CCI	06 43110 20+834147 1687413	##0±2+821	нылжелн <u>с</u> .	січгар 3™сін£ FC8 жн5	•	218 0 J H 2 4 6 J	0645111	01 SOLID COMPONENT (EN/CC)	µMJ=€.87]	янсесяноу	LINEAN Shijini Kirikan
U	2499.88		2.8210				¢	1800.00		7.8/10			
1	734,00	2.256		٥.	υ.		1	1062.00	5.631		b .	۰.	е.
;	728.00		2.9233	150.	758.	758.	5	1052.00	2.031		1445.	144	3.4.5
	489.00		2, 8793	2921.	2345	2395	•	1012.00	1.951		2774	2776.	2716.
~	674.00	2.136		3250.	3214.	3214.	2	1002.00	2.052		e 776.	2776.	2716.
,	659.84		2,8859	3772.	3927.	3927	7	967.44	2.040				
:	629.80	2.072		5206.		****.		937.00	2.046		5767.	- 967.	3767.
jŵ.	384.44	2.026		7666.	7615.	7615.	10	402.0V	2.119		77		7744
14	385,08	8.137		7844	7415.	7615	11		2.107		,,,,,		77
12	349.00	8.117	1./103	10017	#543.	16005	1.5	#71,33 #71,30		7.9255			3374.
11	539.88	2.565		10017.	1.10.0.	11004.	3.4	867,DU	2.107			46.73.	9403
1.	503.00		2.7973	10386.	10178.	101 4.	1.5	867.00 866 86	2.031		9408.	••••3.	9605.
17	494,84	2.0.0		181.54.	1	12141	17	447.40		2.9156	10421.	10.35	10.56.5
1.	472.40	3.044	2 1136	18130.	1214 4	12144.	1.	837.00 417 NB	2.081		11051.	11022.	11072
20	***,80	2.049	••••	13607.		13547.		832,40		2.9252	11235.	11200	11200
**	444,90	2.011		13607.	1	12547.	21	823.40		2.9246	11466.	11477.	116-2
2.0	140.27		2.521	14733.	14706.	1 . 7	25	509.00		2.4043	12/95.	12250	17245.
22	434,00	7.617		150 17.	19005.	15005.	24	807.40	1.951		12384.	1 . 376 .	12178.
24		2.000		14993.	14904.	15005.	25	537.00	5.052	2.9105	12396.	12726.	12:24
51	404,00	2.011		16435.	14.96.	16476	21	772.00	2.052		14106.	1. 1 1 6 .	14010
29	374,00	2.011		17050.	17823.	17023.	20	772.00	2.048		19106.	14010.	14018.
48	344,0U	8.*57		19904.	14277	19277	34	742.00	2.640		13575.	1.4.6.1.	3
31	396,80 814.84	2.416		19504.	14271.	142	34	3 4 2 . DU	á.115		15575.	1++53	1 46 5.
5.5		•••••	2.9210		26701.	20/43.	33	719.44		2.7841	16729.	19689.	1.4417
								709.20		2.7943	17868.	17186.	1/146
								757.00	2.113		17402.	1 304	17504.
							31	672.00	-	2.7241	19190.	14127.	19177
							30	472.00	2.107		19216.	1 . 1	19154.
								668.50		2.0146	19400.	1	191at.
							*1	651,70		2.7805	29283.	20231.	20241
								647.44	2.107		21055.	21014.	2.019
								631.60		2.7841	21535.	21304	21304
								610.00	2.147	2.7201	22454,	22109.	22-48
							• 7	610,00	2.139		22454.	22548	72444
								574,30		2,7613	23534.	22837.	23344
							90	582.44	2.139		25948	2.454	22944
							37	542.00 578.90	5.1.27	2.9136	23948.	25419.	23944
							54	558,90		2.87 5	25257.	P*211.	252 11.
							33	552.00	2.191		25623	29612.	25612.
							56	533.10		2.8882	24543.	26563	26541
							57	522.00	2.046 2.045		27147, 27187	21114.	2711 9 .
							57	317.70	•	2,8623	27349.	27310.	273
							60	492.00	2.008		28552.	28515.	28519.
							62	491.90		2.8059	28348.	20.20	21.3
							6.5	478,90 197,00	2.186	2.9090	24404.	2436	29365
							63	457.40	1.962		30405.	303*20	50352.
							4	452.90		2.8541	30564	30532.	10.15
								418,70		2.8081	82114	32045.	5204
							47	373.70		2.9665	35236.	3+154.	32159
							14	382.04	1.997		33770.	33686.	33446.
							7.8	376.40	-	2.8976	19831.	33945.	13943.
							7.0	852.48	1.997	8,8585	34704.	34413,	\$*413.
							7.8	152.00	1.956		15161	38,72	150-2
							76	847.48 883 34		2,8768	33872.	39:75.	\$5475.
							7.	324.90	3.956		36303.	3.444	31 344
							77	322.40	1.947		36505.	36 196 .	36.596 .
								294,90		2.6745	56859. 17669.	36746.	36786.
								292.00	1.987		37815.	37691	176 11
								272.08 277.92	1.807		57815.	37691.	37691.
								266.40		2,0039	34778.	38/12.	38412.
								262.00	1.807		387 11.	34400.	seene.
							84	249.94		2.8014	39395.	56400. 59251.	38800. 35291.
							87	233.00 232 4 4		2,8643	.0033.	59882 .	39882
							91	252.00	1.012	2,9440	•°v70.	34582. 34519.	-9882. 19919.
							94			2.6210			

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TABLE 6-9 (page 3 of 3 pages). -- Mass of solid in vertical columns of unit cross-section, from BHG-survey data.

			a REMOLE	10°° • 1 *						Strat MOLE	tiņ#~17		
	FLALL	6 - 5	LASATI	CUMULATIVE		140/[4++2)		PLACE	h - ó	UENSITY	CONCLATIV	SAL ID MASS	164/24++2+
J	[[14 D++ 1]	(EN5111 160/011	OF SULID COMPONENT	#HD#2+821	RH0= (4H0)	LINCAR SH, INC	5	D141+	064511+ (64/60)	COMPONENT	#MU=2.021	RHC#KRMO S	LING KG
			(GM/LC)			FOR RHO				(\$8/66)			FUR RHD
6	1446.04		2.8210				101	944,60	2.090		4625.	9758.	\$758.
Ĩ	1140.01		2.8240				104		2.112		9733.	4868.	9068. 10914
2	1138.98	2.20%	2.7262	٥.	٥.	۰.	1	938.15	2.042	2.0903	9955.	1008 .	19047.
1	1136.70	2.1*5		310.	121.	121.	105	956,00	2,039		10043.	101 .	10193.
- 2	1100.60	2.116		233.	234.	238.	10/	926.14	2.033		10546.	10668.	19668.
1	1194.24		2.7149	r40. 300.	245.	307.	104	923.11	2.016	2.9191	10701.	10499.	18699.
•	1135.45	2.872		367.	375.	375.	110	970,96	2.024		10793.	10411.	39911,
34	1191.04	2.00/		577	539.	319.	114	\$12,20		2,8948	11243.	11353	11393.
18	1126.84	2,145		645.	677.		11.2	989.78	2,039		11364.	11473.	11473.
14	1123.60	2.144		760.	776.	116.	112		2.044		11791.	11834	11834.
i i	1183.93	2.112		\$27.	ANS.	443. 113	110	846.43 878.WV	2.048		12043.	12132.	17157.
17	1181.74	2.087		*14.	+ 35	***	11.	889.00		2.9103	12.23	12.11.	12514.
1.4	1128.30		5 30.	418.		1066.	154	840.7/	2,544		12837.	12921.	12421.
29	1117.90	2.022		1126.	1151.	1191.	151	\$76.44	2,197		3 30 7 8	15150.	11144
72	1119.24	2.0**		14*4.	1257	1257.	123	\$70.20	2.1.30	5.9109	13309.	14462.	13462.
2.8	1113.96	2.047		13.6.	1 342.	13.2.	124	867.22	2.071		13536.	13607.	13668.
25	1112.70	2,002		1921.	1 50	1*50.	120	841.43	.010		13#03.	13864.	13867.
25	3339,99	2.561		1443.	1472.	3472.	121	837,32 832.60	2.13*	2 9296	14016.	14078.	34078.
20	1189,49	2.866	2.7303	1507.	1937.	1537.	129	852.60	2,154	•••••	14267.	14324.	1+32+.
29	1107.54	2.328		1617.	1649.	1647.	130	850,00	2.041		143.7.	14412.	14417.
36	1104.96	2.94		1756.	1793,	1793.	138		2.130		146AD.	14729.	14724.
38	1101.67	2.196		1430.	3867.	1867.	134	438.41	2.057	5.9292	1	15031.	15031.
3.9	1148.20	*****	2.7544	1917.	1950.	1993.	1.15		2.085		19160.	15199.	19194.
33	1101.04	2,159		1972.	2011.	2011.	336	832.82	2,125		15449.	15308.	13443.
51	1099.47	2.106		2044.	2104.	2104.	130	818.51	2.067		19993.	14526.	19376.
50 37	1098.00	2.104		2130.	2172.	2242.	3.87	825.20	2,085	2.96.92	15796.	15823.	15423.
	1092.60		2.7293	2408.	2496.	2456.	244		2,0=6		10207.	16225.	16321
	1072.30	2.071		2456.	2505.	2363.	1.1	810.47	2.002		16044.	16809.	16404.
**	1007.77	2.076		2651, 2788	2784.	2784.	1.44	807,03	2,022		16356.	16568.	16368.
	1803.04	2.11#		2095.	2953.	2753.	198		2.128		16796.	14804.	16804.
- 25	1079.50	2.088	2.72*2	3072.	5087.	3047.	148	797.57	2,135		17048.	10701.	17091.
- 18	1073.00	1.967		3361.		3979.	117	795.20		2,9246	17221.	17721.	17221.
	1065.7V	1,481	2,7242	3729.	3804.	3024	131	786,24	2,166		17504.	17499	37499.
51	1041.34	2.010		3914.	3993.	3993.	ise	786.57	2,088		17414	17607.	17607.
54	1033.78	1,948		4173.	4257.	4257	134	144.61	2.036		17700	17691	17691.
5.	1052.94	2.014		4334.		4401.	155	781,23 779 31	2.017		17865.	17853.	1791
56	1049.74	1	2.1242	**6*.			157	775.85	2.044		1#136.	18120.	18120.
51	1048.64	1.945		4312.	4603.	4403.	150	769.14	2.026		18-52.	16222.	10422.
59	10-4.84	2.034		*/11.			164	767,90	2.110		18518.	18495.	19495.
6.	1045.00	2.067	2.73*3			4935	164	764 .87	2.123		18678	18452.	10452.
64	1033.44		2.7301	5161.	3265.	5264 .	365	762.50		3.4500	10014.	18786.	18786.
	1026.49	1.800		5977.	5647.	3487.	160	799,74	2.135		14750.	14718.	18918.
49	1023.44	1,950	8.1291	****	5724.	5795.	267	752,84	2,119		19312.	14046.	19274
- 52	1022.05	1.987		5727.	4642.	56.2.	160	751.50		2,9190	19381.	19342.	19342.
	1019.04	1.090		5841.		1978.	170	247.14	1.472		19559.	19516.	19516.
74	1018.03	1.00%	8.7120	5786. 6140.	6108. 6259.	6106. 6254.	174	7 40.0 4 787.84	5.080	8,9183	14889. 28036.	19848.	17840.
74	1011.67	1.941		4205.	6331	683).	170	783.60	2.007		201.0	20087.	20067
7.	1004.91	2.149		6267. 6960.	4579.	6599.	173	726.20	2.047		20606.	20345.	20345.
73	1074.79	2.146		4259.	6692.	6692.	176	725.14		2.8036	20725.	20663.	20663.
- 11	1004.70	2.140	8,1742	6467	6805.	4403.	170	714.37	2.080		21184,	21115.	21115
78	1000.44	2.157		6811.	6950.	6750.	179	714.14	2.050		21205.	21137.	21137.
	994.98	1.976		7056.	7292.	7202.	181	791.70	1.007		21792.	21716.	21716.
81	993,20 992-75		2.7323	7137.	7202.	7282.	152	700,20 699.95	2.054	2,6948	21864.	21787. 21818.	21010.
	998.90		2,8633	7248.	7393.	7345	1	676.11	2.045		22065	21 485	21985.
84	989,32 987,34	2.071 2.071		7326.	7471. 7975.	7575.	140	678.23	2.02		22106.	27208.	22248
	***.18	2.014		1975.	7717.	7717.	147	668.51	2.002		22434	27149	22349.
	\$75.44	8.045	2.8740	Av22.	997C. 8161.	8141.	167	680,20	2.050		22839	77748	22740
87	973,91	2.041		#112.				678.US	2.057		22945. shu to	22852.	22832.
		8.010			.445.		1.44	670.Š.	2.106			21230.	23290.
74	962.94 961.37	2.235	8.7848	8614. 8699.	#751. ##37.		194	646.8U 684.47	2.034	2.4928	28506. 28659.	23527.	23527,
	\$78,86	2.299			\$017.	9017.	195	660.03	2.05		28747.	23693,	23693.
**	933,85 938,84	8.197 8.188		9059. 9198.	4140.	4505	144	626,83	2.073		24014	23421.	23907.
* /	958,73 830-84	8.140		9189. 8308	9827.	9887.	170	693.11	2.367	2.8764	24383. 24284	2.077.	24877. 24847.
-		1/1-1					111	479.11			24.237	24.224 .	24226.
104	441,74	5,334		7489.	962£.	7626.							

SDREMOLE NON-17

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TABLE 6-10. -- Mass of solid in vertical columns of unit cross-section, from Y-Y survey data. Table continued on succeeding 19 pages.

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			SURF HOLE	00P-17						BUREMOLE	008-1-		
	PLALL	6 - 6	DENSITY	CUMULATIVE				PLALL	6 - 6	0[45177	CORD, A FLVF	10: 10 HASS	
•	08674 (1887	0EN5111 16#/101	07 80636 (Seponent (Sevee)	AHCE7.821	ныресямоу	1147ав 5Р[140 F08 8нΩ	J	01111	0145'TT (64/50)	NE BOLID COMPONENT COMPLES		RH()=(BHD)	, 13,132 3P(14) 108 सम
291	447,58	\$1014			24352.	24352.				2.6246	38266.	18205.	38205.
202	6+4.3/ 643 3/	2,702		24605. 24706	24492.	24492.	80 a	339.41	1.986			5842*.	18-24
-0.	649.44	974		2.004.	24691.	2***1	404	554.7	1.958		36620.	24417.	38614.
205	637,64	2.082		24925.	26809.	24889	345	5.31.74	2.002		58758.	14755.	34744
107	458.11	2.947		25111.	2	2	407	520.60	1	2.8345	38915	34907.	39969
208	633.04	2.041		25417.	24048.	25048.	300	3.3.17	2.010		39136.	34131.	39351.
114	628.50		2.8508	29376.	24244	7*7*9	31.4	319.41	1,972		37847	39314.	\$9351.
214	647,77	2.007		29404.	25245.	27287.	514	316.06	1.474			SARAF.	3**Ar .
213	663.46	2.036		25t 28.	25507	25507	313	313.**	1.950		34014.	3470'.	1960.
514	622.10	2.000		23467.	2.347.	25547.	31"	311.74	1.894			39676.	
-15	615.41	1	2,8610	275.44	25711.	25711.	315	105.74	1,939		34742.	3078°. 10995	
	611.46	1. ** *		25868.	24744.	*****	817	401.44	2.007			47367.	4916F .
21.	631, ** 639, 37			26433.	26100.	74104	917	247.51	2.061		+C.U.	40.01	4620
1.1	603.04	5.111		26*53.	26324.	26374	520	207.50					-0.561
224	493.74	2.942	2	26676	26343.	28341	526	299.13	3.950		4044C	40446. 40528	46446.
423	599.51	6.834		24785.	26450.	26456	825	244.44	1, 114		+038e.		45.44
223		2.078		27103.	26969.	24764	323	284,83	1.938 1.984		+C719.	40116.	40716. MCALL
226	587.44	2.0**		27381.	27240.	2 . 5 . 6	3.24	286.44		2.7467		40002	408A
221	589.33	3.050	5.6243	27528.	27297.	27387	32/	285.81	1.919		-0987.	40 48 P.	
229	582.1*	2.011		27635.	27494.	27444	\$24	279.00	1.832		*11	41170.	-1116
230	980,UJ 478,50	3.02+	2 7383	27762.	27602.	27602.	290	217.17	1.886		41276.	1221	41279.
234	554.74		2,72+2	28795.	20675.	284 * 3	334	214.78	3.907	1.1311	.1.5.2		
234	552.00	2.054		29119.	24004.	29096.	834	273.94	1.067			* 1 9 7 2	41342.
235	333.60	1	2.7343	29934.	29842.	29842	335	268.76	2.026		41427.	41629	41545.
236	545.40		2,7564	30279.	\$0190,	5C197 .	3 5 8	266.18	2.005		.1749.	41751.	41751
830	502.60		2,7242	31 21 2.	31241.		337	264.0V 264.UV	1.971	2. = 3 < 1	*1000.	41830. 41815	4181C. 41815.
237	502.20		2,7365	31429.	31254.	51254,	837	260.60	1.967			. +0944	
24.	441.20		2.12.2	32259.	32204.	32204	341	255.43	1.979		*2100.	42102.	42102.
	4'0,20		2,1202	52747.	32709.	3,705.	344	255.40		2.8454	.7 . 1 .	4 2 2 5 9	*225*
244	434.24	1.410	2.7245	33366.	33337.	33137	345	253,14	1.993		*****	w2315.	*231**
542	\$35.37	1.435		33403.	33375.	\$3375.	845	250.40	2.005		.2.75		
54P	****	1,03		33637	33452.	33432	346	243.35	1.972			.26*1.	42641.
248	648.34	1		33717.	\$ 3716.	32776	3	291.45			.2891		
1	446.15	1.44		33797.	33778.	93778	447	238.47	1.964		43009.		
<. 1	441	1. 458		34020.	34014	3+01+.	351	754 80	1.990		43188.	* 302* .	43192.
214	441,40		5.1101	34046.	34031.	34031.	158	253.14		2.8625	43234	43232.	432.52
254				3*##1	34201.	34201	494	220,70	1.971		*3557.	· · · · · · · · · · · · · · · · · · ·	43337,
233	438.87	1.441		39 446 .	34337.	\$4337.	835	226.63	2.019		43957	41551	43551.
231	429,99	1.990		34527	3.518.	1.510	\$57	222.85	7.048	2.4492	43746	* 1633.	43433
528	429.13	1.92/		39365.	34 55 7 .	3.3.1	854		7.016		. 3731	* 1971.	* 3 * 21 .
460	446.40	1	2,8277	34691.	34683,	34641	354	214,27	1.929		#*126. #*271	**115. **26*	44115.
261	941,84	1.841		34875.	3+868.	34868	361	201.20	•••••	2.8633		4 . 4 36	*** 56
263		1.027	8.7739	34931.	34929.	34924	362	206.00	1.913				
26.4	416.66	3.791		\$2071.	3=044.	13066	36*	193.10	1.948		. 5079		
265	412.30	1.878		35121.	35116.	27116,	360	192.20	7.	2,6299	45120.		\$\$105.
267	411.70		2,728;	35273.	34272.	3.272	35.1	187.44	1.938		45313.	45295.	1295
26.	411.07	1.910		35961.	34300.	35300.	360	186.50		2.8341	*5585.	44.367	43367.
270		3.984		33545.	35548.	3***8.	570	183.85	2.010				45465
272	404.17	1.946		33003.	35600.	33488.	574	101.74	2.081			45367.	43567.
275	68D, 7V		2,7426	33754.	35768.	33765	37.5	177 30		2.8548	45798		
274	377.46 878.67	1.050		35011.	35818. 18858.	33818.	374	176.38	2.125		43858		
176	396.80	2.810		33930.	39939.	35759.		171.17	2.002		46172	******	46100.
279	300.13	1.988		36701.	36312.	36312	371	367.95	5.054			44102.	
	384,75	1.477		34292.	34403.	\$6403.	579	144.20	1	2.8041		46438.	46410.
280	384,41 188 4/	1.414		36482.	36493.	36445.	58U	141.48	1.936		46371	16390	*4550.
284	\$40,11	1.4**		36866.	36679.		344	156.70	1.969			46 387.	*****
28.4	377,96	1.874	7 7567	34749.	36763.	34763	38.4	157,90		2.8233		*66*7.	****
	378.41	1,851		36933,	34848.	34844	383	124.44	1.933		46833.	*683*.	46834.
284	374,93	1.932		56847.	34902,	36902.	386	131.14		2,8745		44 458.	46938.
284	372.04	1.958		37035.	37053.	170 53	380	193.29	1.015	2 85/1	.7010. .7865	*6986.	46786.
284	368.71	1.915		37348.	37166,	37166.	387	142.57	1.625		.7337	*****	47311
290	365.07	1.967		37204.	37225, 37317.	37223.	394	135.42	1.777		. 7577.		• 7550 .
274	363,37	1,951		375*3.	37342.	37391	392	1 29 . 47	1.905		.7010	47740.	47740
294	358.62	1.007		379=6.	37463,	17961.	393	128.50		2,8165	. 784 1		
295	350.17	1.920		37589	17605.	\$7405	193	12	1.430	2.9259	47785.	4743°.	
295	896,04 384,20	1.917	8.9144	37681.	37696.	37496.	396	153.84	1.075				48027.
298	\$52.17	1.001		378+3.	37855	37855	398	120,01	1,922		48216.	48147.	48347. 48382.
277 277	350,04	1.450		37935.	37943,	\$7943. \$817-	397	110.27	1.927			*1255.	48235
				3-107.	361,3	34113.	-00	111.44	1,767		48325.	N#292.	48292.
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			BOALHOLD	008-17						\$ 38EH01 L	04 8+2 1		
	PLACL	6 - 6	DENSITY	CUMULATIVE	301 10 MASS	(bP/CR4+2)		PEACL	5 . 5	DENSITY	COMULATIVE	511 11 MASS	14475 *****
	0LC14 01818	(64/CC)	OF SULTO COMPONENT	#HD+8.821	RHCHCRHO>	SPLINE	J	(1/1)	DENSI1.	OF SULID	#HU12.421	REGREAMUS	1311 AP
	-		IBH/LC)			FOR AND				(SH/LC)			FUS BHC
481	113.11	1,936			N& 567.	48347.	U	3800.00		2.0210			
	114.80	1	2. 9275			44444,		452.70		2.8432			
***	111.41	1.717			46551.	4835 1.	4	421.50		2.7202	-		
485	100.03	1.405	2.9290	+#771	48726.	+8726.	;	*11.4V	3.938	2.2144	76.	70	С. ТС.
	194.47	1.919			48765.	.4765.	•	411.0/	1.989		84.		
	108.01	1,931		49379.	40711.	47125.	<u>'</u>	408.72	1.50%		164.	164.	184.
414		1,901			*****		•	404,41	1.741		3	345.	345
*11	79,44	1.929		*****	49267 ,	49267.	10	402.85	1.713		397.	398.	398.
		1.430	2.9892				18	401.24	1.748	2.7160	· 5r		
* 1 *	63.90	1.074		49766.	**701.	**701.	1.2	399.10	1.693		202.	3.44	· · · ·
414		1.050			49857.	49857.	1.5	396.71	1.725		540.	543.	543.
417	78,33	1.878		49997.	49928.	49927.	1.	394.61	1.759		673.	6 . 7 .	477.
414	79,79	1,004		40149.	50084.	50086.	1.	394.23	1.741		284	691. 789	691. 789.
424	12.14	90.5		50231.	50:57.	50157.	1.	349,54	1.697		Aut .		851.
421	66.90		2,9220				20	388.36	1,716		885,	A90.	840.
423	36.27		2.8210				24	585.25	1.643		\$83.		
424			2.4210				23	383.21	1.660		1042.	.048.	10.8.
							25	318,70	1.865		1197.	1148.	1199.
							26	377.34	1.876		1<56.	1.62.	1242.
							28	373.46	1.745		1303.	1310.	1310.
							54	5-1.10	1.748		1.87.	1443	1441
							30	370,34	1.762		1515.	1921.	1521.
								568.3"	1.886		1393.	1399.	1599.
							3.2	366,40	1.006		1673.	1680.	1680.
							35	363.00		2.6307	1013.	1819.	1019.
							35	362,51	1.676		1835,	1838.	1838.
							3.0	339,38	1,911		1965.	1871.	1970.
							37	357.01	1.873		2430.	2035.	2035.
								354,27			\$169.	2173.	2173.
								\$92,72	1.754		7227.	\$2.51.	2751.
								399.20	1.672		2244.	2297.	2297.
							• 5	544.80		2.83*3	2422.	2425.	2 2 2 3 .
								346,06	1.731		2447.	2450.	2.30.
								3=3,3,	1 782		2543.	2545.	2565
							**	342.1:	1.629		2587.	2589.	2549.
							51	357.45	1.841		2773.		2174
							54	336.67	1.876		2805.	2805	280*
								334.73	1.006	8.8248	2882.	2832.	2032.
							53	333.94	1.798		2911.	2910.	2910.
							57	331.14	1.755		3009.	3007.	30-7.
							58	329.44	1.751		5015	. 74	1074
							39	327.27	1,804		3346.	4144.	3144.
							61	324.94	1.004		3240.	3238	3738.
							54	522,96	1.881		3321.	3318.	3314.
								320.90	1.00-	2,8345	3404.	3365.	3365.
							63	319.04	1.408		3482.		34 78 .
							61	517.87 815.91	1,900		353). 3612.	3577.	3597. 8608.
								\$15.75	1.820		3658.	3633.	3453.
							74	313.17 309.64	1.813		3717.	3713.	3715.
							71	304.90	1.815		8967.	3962.	3962.
							72	303,34	1.80%	2.8.74	4826.	4020.	4020.
							74	105. **	1.617		4085.		+074.
							75	301.01			4165.	-156.	158.
								297.50	1.889		4352.	4345.	4845.
							78	294,76	1.80%		**65,		4459.
							50	291.00	3.762	2. 718	4626.	4621.	4621.
							84	290,45	1.942		4632.	.627.	4677.
							63	288.87	1,925		4717	4677.	4713.
							87	267.52	1.010				
							85	283.41	1.798		4898. 4957.		4896.
							<u>•</u>	281.06	1		5024.	5022.	5022.
							84 84	279,99	1.935		5093.	5892. 8188	9042. 5144
							94	275.51	1.995		5267.	5267.	5247
							91	275.10	1.899		5286.	5203.	5285.
							93	271.65	1.424		54 26.	3476.	5476
							9 4 4 3	278,70	1.861	2.81/4	5463,	5465.	2.61.
							96	265.74	1.861		3668.	5660.	3460.
								264,61	1.852		5706.	9786.	3764.
								261.40	1.873		5846.	4737, 5845.	5845.
							100	260.54		2,4633	5078,	5869.	5869

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in the support of a

				BOOLHULE	05#-21									
		PEACE	5 - 5	DEWSITY	CUMULATIVE	30. ID #855	158/18442				BIREHOLD	0+1-20		
	5	C IF T I	DE45311 (6+/CC)	0" 50110 Cuaronent		B	LINFAR		L DIPT-	1 - 5 ntusti	ULNSATA DE SULT	CUNULATIN	C FOLIC HAS	+ + + 2 -
				168/10)	1-012 -21	440¥C#MC)	FOR RHC		0111	(CONTINEN.	H-1-1-1-42	A RUDECPHI	a lafar Sfi far
							-				tem/cc:			flo Ant
	104	260.3V 259.13	1,861		5878.	×877.	50		e 1406.00	u l				
	193	255.77	1.887		60.4	5922.	\$922. 60×6		1		2.0210			
	107	252.00	1.920		6132.	6129.	6179.		* 571.44		2.867#			
	196	230.31	1,779		6272.	4267	\$203.		\$72.40	, ,,,,,,,,,,,,,,,,,,,,,,,,,,,,,,,,,,,,,	2.7473	۲. ٥.	D .	¢.
	10.	249.10	1.782	8.8508	6294.	6789.	6289.		570.	1.969		5.		•••
	1.79	247.58	1.826		6386	6352. . 181	\$ 392.		348.14	1 1 1 1		197	108.	1-1
	111	245,05	1.903			6475.			· 967,31 · 569,98	1 1.448		278.	25.	232.
	114	236.90	1.908		6827.	4638.	6638.	1	V 963.70	1.910		512.	· · · ·	
	113	238,32	1.915		6865.		6812	1	1 562.61	1. •1 •				
	115	230,15	1.451		7025.	1015.	7013.	1	3 559,UB	1.847				514.
	110	228.58	1,845		7166.	7156.	7156.	1	• • • • • • • • • • • • • • • • • • •	2.000		1.	726	726
	11.0	225,V6	1.405		7182. 7800.	7171.	7171.	i	• 555.eu	1,484		801.	A29.	• 2 ° .
	120	218 57	1.826		7589.	7377	, , , ,	1	7 <u>551</u> 90 5 51155		2.7.42		917	
	121	218,90	1	2,0921	7515.	- 502.	7502.	1	* ***.**	1. •••		• 1 • .		• •
	124	216,83	1.068		1625.	615	7615.	2	0 548,09 1 517 <i>1</i> /	1.4.5		i	1010.	1.14
	174	212. 4	1.970		7741.	,,,,,	7727.	2		1.85.			1104.	1104
	120	211.74 209.NU	1.998		7847.	7931.	10.51	8. 29	3 <u>561</u> ,84	1.45-		1331		
	15.	208.70	••••	2.4631	7994.	7456.	7956.	7		1.912		1947.	1 .	1 . 1
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	1 4 0	20. 75	1,920		6147. A244	A12.	8129.	2	4.57.44	1. •1•	2.1282		1	
	114	149.60	1,734			4346.	C 3 6 6 .	2.	535.76	1.951			1614	1563.
	120		1.935		**36,	-417.	** 17.	5 1	55	1		1645.	16	
	194	193.75	1,857		4639	4619.	8414.		530.81	1. 114		17.4	1725.	1429
	136	1 40 . 20	1.798		#733, #738	.713.	.715	,,,,,,,,,,,,,,,,,,,,,,,,,,,,,,,,,,,,,,,	521,44	1.994	1 1200	1828	1863.	101
	137	190,00		2.8341	8786	A758. A765.	8758.	33	327.34	1.921		1945.	1911	19.0
	1 1 4	196.29	1.017		4047.			, ,	524 44	1.1.1		1.445	2033.	2043
	140	183,55	1.923		9038	9017	1017		141.44	1.875		2078	-144	20mm.
10.10 1.013 1.013 1.010 <td< td=""><td>142</td><td>178.06</td><td>1,471</td><td></td><td>9118. 9265</td><td>4078</td><td>90 en .</td><td></td><td>510,33</td><td>1.060</td><td></td><td>21 91.</td><td>2:35.</td><td>273</td></td<>	142	178.06	1,471		9118. 9265	4078	90 en .		510,33	1.060		21 91.	2:35.	273
No. 1.1.1.1.1.1.1.1.1.1.1.1.1.1.1.1.1.1.1.		176.10	1,813		9412.	4241			5:7.15	1. * 50		< 376 .	2169.	2364.
111.00 1.01.00 2.010 2.010 2.010 2.010 111.00 1.000 000 000 000 000 000 000 111.00 1.000 000		174.95	1,982	*. *]##	4516. 4847	4245.	\$2.5.		5 5			2412	• • 8	
<pre>Not 100 00 100 100 100 100 100 100 100 100</pre>		178.97	1.913		** 1 7		7316		111 60		2,7160	4.0	2672	2694.
10.10.10.10.10.10.10.10.10.10.10.10.10.1		169.06	1,8+1		7745	• • • • • •	**17.	**	904,94	1.764		22.15		26.97
<pre>1, 1, 1, 1, 1, 1, 1, 1, 1, 1, 1, 1, 1, 1</pre>	50	161.10	1.837		9672	4651	9451		506.27	1. ** :		24.1	2 1 7	2827.
30 100, 0 100, 0 100, 0 1.00, 0 <td< td=""><td>1</td><td>142.01</td><td>3.798</td><td></td><td>9794. 9848</td><td>• 7 7 3 .</td><td>9775.</td><td></td><td>503.**</td><td>1.939</td><td></td><td>2005</td><td>2953.</td><td>2955.</td></td<>	1	142.01	3.798		9794. 9848	• 7 7 3 .	9775.		503.**	1.939		2005	2953.	2955.
30 33 <td< td=""><td>52</td><td>156.51</td><td>1.470</td><td></td><td>****</td><td>9975.</td><td>9975.</td><td>50</td><td>501,0' 900 by</td><td>1. * 57</td><td></td><td>30.4</td><td>3130.</td><td>3060.</td></td<>	52	156.51	1.470		****	9975.	9975.	50	501,0' 900 by	1. * 57		30.4	3130.	3060.
30 132, 64, 2, 047 10007 10007 10003 30 90, 07 1, 077 332, 5 1007 30 133, 67 1, 977 10007 10003 30 90, 07 1, 045 335, 7 1007 30 133, 67 1, 977 10007 10000 10000 300, 7 1, 977 335, 7 1007 31 1, 977 10000 10000 10000 10000 300, 7 10000	5.	138.39	2.042		20090.	10069.	19969.	54	497.95	1.830		3171.	3105.	2183
37 10000 10000 10000 30	39	192.61	2.047		18303.	10201.	10201	53	**5,**	1.877		3312	1578	3200.
30 140.30 1.9640 100.00	57	149.47	1.992		18378.	10356.	10356	55	492.86	2.046		3375.	1443.	
av ia ia <td< td=""><td>59</td><td>148.30</td><td>1.964</td><td></td><td>10504.</td><td>10-82.</td><td>10482</td><td>31</td><td>486,70</td><td>2.008</td><td></td><td>3536</td><td></td><td>3515.</td></td<>	59	148.30	1.964		10504.	10-82.	10482	31	486,70	2.008		3536		3515.
11 12 13 12 10 <td< td=""><td></td><td>144.78</td><td>1,955</td><td>4,8187</td><td>19687.</td><td>1050%.</td><td>10544</td><td>5.</td><td>446.37</td><td>2.079</td><td></td><td>3451.</td><td>.725.</td><td></td></td<>		144.78	1,955	4,8187	19687.	1050%.	10544	5.	446.37	2.079		3451.	.725.	
14 14 14 14 12 12 14 12 12 14 12 12 14 14 12 14 14 12 14 <td< td=""><td>42</td><td>142.UJ 159.66</td><td>1.839</td><td></td><td>39776.</td><td>10752.</td><td>10752</td><td></td><td>485,02</td><td>2.0%</td><td></td><td>3832</td><td></td><td>3964</td></td<>	42	142.UJ 159.66	1.839		39776.	10752.	10752		485,02	2.0%		3832		3964
131.00 2.73%2 390. 000.00 2.73%2 390. 000.00 2.73%2 390. 000.00 2.73%2 390. 000.00 2.73%2 390. 000.00 2.73%2 390. 000.00 2.73%2 390. 000.00 2.73%2 390. 000.00 2.73%2 390. 000.00 2.73%2 390.00 000.00 2.73%2 390.00 000.00 2.73%2 390.00 000.00 2.73%2 390.00 000.00 2.73%2 390.00 000.00 2.73%2 390.00 000.00 2.73%2 390.00 000.00 2.73%2 390.00 000.00 2.73%2 390.00 000.00 400.00 300.00 400.00 300.00 400.00 300.00 400.00 300.00 400.00 300.00 400.00 <td></td> <td>136,94</td> <td>1.071</td> <td></td> <td>10844.</td> <td>10841.</td> <td>10841.</td> <td>*1</td> <td>481.87</td> <td>2.035</td> <td></td> <td>3940. 3985</td> <td>****</td> <td>3969.</td>		136,94	1.071		10844.	10841.	10841.	*1	481.87	2.035		3940. 3985	****	3969.
de 131.00 2.9866 11200. 11171. 11371. de 171.00 002. 1400. 11200. 11207. 11207. 11277. 11277. 11277. 1277.		131.82	1,435		11477	11045.	110.9		480,42	2.960	2,12*2	5999	ORC.	-09C
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87 40 1.000 00000 0000 0000 <	6.	115.00	1.415	2.1254	11887.	11287	11777.		.7	2.075		4278.	. 36 .	4365
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72 82 82 82 82 873 4375 4375 4375 76 850 76 1993 4272 4375 4375 76 850 76 1993 4272 4375 4375 76 850 1.998 4272 3075 4375 4375 4375 76 850 1.998 3172 3275								70	N64,80	• • • •	2.7201		4714.	*716
$\begin{array}{cccccccccccccccccccccccccccccccccccc$								74	461.90	1.953		*738.		4855
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$\begin{array}{cccccccccccccccccccccccccccccccccccc$									434,83	1.975		5176	5282.	2227
77 a = 0, 10 1, 90. 50.0 50.0 80 a = 0, 10 1, 90. 50.0 50.0 84 a = 0, 10 2, 01.1 350.1 50.0 84 a = 0, 27 2, 01.1 350.1 50.0 85 a = 0, 27 2, 01.1 350.1 50.0 85 a = 0, 27 2, 12.1 37.0 40.0 85 a = 0, 27 1, 91.2 37.0 40.0 85 a = 0, 27 1, 91.2 37.0 40.0 85 a = 0, 27 1, 91.2 37.0 40.0 85 a = 0, 27 1, 91.2 37.0 40.0 86 a = 0, 27 1, 91.2 37.0 40.0 86 a = 0, 27 1, 92.0 39.0 40.0 86 a = 0, 20 39.0 39.0 40.0 87 a = 0, 20 1, 97.0 40.0 40.0 87 a = 0, 20 1, 97.0 40.0 40.0 97 487.0 1, 97.0 41.0 42.0 47.0 9								78	458.91	1.95-		5245	5351	1 1 4 4
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#3 #3 #3 #3 #3 #3 #3 #4 <td< td=""><td></td><td></td><td></td><td></td><td></td><td></td><td></td><td></td><td>**1.72</td><td>1.911</td><td>4 7171</td><td>5731.</td><td>3844</td><td></td></td<>									**1.72	1.911	4 7171	5731.	3844	
07 4.54.04 1.954 3.671, 4.962 08 4.36.04 1.954 3.951, 6.660, 08 4.36.04 1.951, 3.961, 6.660, 09 4.36.04 1.951, 6.960, 6.180, 10 4.32,140 2.9140, 6.960, 6.180, 11 4.32,140 2.9140, 6.186, 6.100, 12 4.32,140 2.9140, 6.186, 6.100, 14 4.32,140 2.9140, 6.186, 6.100, 14 4.32,140 2.9140, 6.186, 6.100, 14 4.32,140 2.9140, 6.186, 6.100, 15 4.22,140 2.9140, 6.186, 6.100, 16 4.92,140, 1.978, 6.186, 6.421, 17 4.92,160, 1.978, 6.412, 4.177, 16 4.92,160, 1.978, 6.412, 4.172, 17 4.92,160, 1.978, 6.421, 4.172, 19 4.92,160, 1.978, 6.421, 4.172, <td></td> <td></td> <td></td> <td></td> <td></td> <td></td> <td></td> <td>85</td> <td>439,96 489.37</td> <td>1</td> <td></td> <td>58.54</td> <td>5975</td> <td>2007.</td>								85	439,96 489.37	1		58.54	5975	2007.
09 436,82 1,900 3741, 6.041, 09 435,42 1,903 5044, 6.184, 90 432,100 1,873 6.044, 6.190, 91 432,100 2,1140 6.184, 6.190, 94 430,55 1,874 6.194, 5.374, 94 430,55 1,874 6.194, 5.374, 94 430,55 1,874 6.194, 5.374, 94 430,55 1,874 6.194, 5.374, 94 42,04 1,976 6.147, 4.177, 95 425,46 3.005 6.147, 4.177, 95 425,46 3.005 6.147, 4.177, 95 425,46 3.005 6.146,4621, 4.922, 97 423,87 3.076 6.922, 6.458,4621, 97 423,87 3.076 6.922, 6.458,4621,								67		1.956		5871.		****
1 1.753 6084, 1.803 1 1.875 6184, 1.870 1 1.871 2.110 6184, 1 1.871 2.110 6194, 1 1.871 1.871 2.110 1 1.872,10 1.871 2.110 1 1.870,00 1.870 6194, 1 1.870,00 1.870 6194, 1 1.870,00 1.870 6194, 1 1.870,00 1.870 6484, 1 1.870,00 6484, 6421, 1 1.870,00 6484, 6421, 1 1.870,00 6484, 6421, 1 1.870,00 6484, 6421, 1 1.870,00 6484, 6421, 1 1.870,00 6484, 6421, 1 1.870,00 6484, 6421, 1 1.870,00 6484, 6421,								69	436,02	1.980			6064. KIIA.	6066. 4115
91 92,10 9184,6 6200, 94 930,55 1,874 6199, 8376, 94 930,55 1,874 6482, 4350, 94 930,55 1,874 6482, 4350, 94 947,04 1,976 6482, 4370, 95 923,69 2,001 6986, 6421, 95 923,69 2,000 6986, 6421, 97 923,89 3,070 6932, 6322,								yu	4 5 2 . VU	1.895		6064.	6190.	6190
94 428,54 1.844 4265, 4.547, 94 428,54 1.846 4265, 4.547, 95 427,04 1.978 4.417, 4.546, 95 428,44 2.005 4.846, 4.642, 96 428,44 2.000 4.8484, 4.622, 97 428,44 2.000 4.822, 4.638,									432.14	1 87-	2. 11+0	6199	# 37 8 .	6294 . 632 P
9 927,03 1,998 54.2. 4.77. 9 925,98 3,205 64.14. 4.946, 9 9 928,99 3,000 64984, 4.621, 9 9 928,99 3,076 6922. 4.638,										1.478		6262	4 3 4 3	6341
96 828.68 2.000 6522. 6636. 6621. 97 828,89 3.000 6522. 6636.								*5	427.0 <i>8</i>	1.978		6-14.	6346.	6346
97 423,89 3,976 9322° 4458.									424.60	2.000		6486. 6523	4421.	6621
98 421.95 i sui 4093, 4445.								**	428,89 421,95	1,976			6693.	663F
••••••••••••••••••••••••••••••••••••••								-	*18.*4	2.000		6626. 6836	4785	6785.
100 m1/,64 2.014 6485, 6485,								100	-17,82	5.014				4944

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			BORLHOLE	081-20						BOREHOLE	081-20		
	PLACL		DENSITY	CUMULATIVE	SOL SD MASS	15#/(#++2)		PEACL	6 · ·	DENSITT	CUMULATIVE	10,10 ++55	167/5=++21
	36P1*	0EN\$111 (\$4/001	DF SOLID COMPONENT	440s2.871	-	110EAR 3Pi19E	J	DLFTI	DENBIT: (6x/CC)	DF SULID COMPONENT	AMU#2.82;		LINCAR SPLINF
			154/661			FOR RHD		•••••		150/651			FOR 840
101	416.90	2 846	5.1344	6878. 7169.	7020.	7812.	202	268.66	1.949		10622.	18697.	18697.
10.	.07.5.	2.017		7534.	7483.	7483.	28.4	161.10		2.9198	19835	10976.	18906.
104	*05.#/	2.027		7407.	7544.	7616.	284	360,27	2,879		19961	10932.	19952.
104	.02.15	2.057		7563.	7712.	7712.	204	156.65	2.079		19961.	19124	19129.
101	401.44		2.8146	768%.	7754.	7754.	201	198.91	1.953		19191.	19257.	19257.
100	399,21	2.093		1841.	7990.	7990.	222	191.77	1.942		19844.	14342.	19926.
110	396.47	2.074		7901.		8050.	81.0	148.98	1,997		19435.	29445	19995.
111	391.37	2,005		8130.	8277.	8412.	214	186.60		2,9297	19518.	14577.	19577.
114	387.70	2.02	c 1974	8304.			21.4	144.11	1.997		19631.	14688.	19685.
11.	385.00	2.047		0392.	8936.	8536.	\$1.4	1.0.90	1.924		19772.	14826.	19826.
112	381.96	2.02*		8782.	4436	4436.	213	132.85	1.17		19938.	19941.	19941.
117	\$77.44	2.042		8811.			117	135.00	1.885		19987.	20137.	20037.
110	375.69	5.05P		8867.	4026.	9126.	210	132.75	1.692		20116.	20164	20144.
120	3/2.99	2.041		4095.		9233.	224	131.94	1.750	2.9096	20151.	71200	20200.
184	349, 42	2.034		7197.	\$527.	9327.	221	131.89		2.8216	20153.	20200,	20200.
124	367.40		2.8341	7287.	4424.		222	124.05	1,730		20425.	20372.	20572.
124	363.31			9379.	9515.	4515.	22*	175.74	1.951		20116.	20463.	20363.
127	362.76	2.044		9513.	-647.	9649.	223	119.02	1.917		25687.	2-714.	20794.
121	359.44	. 059		7665.	. 5540	9822.	420			2.0716			
124	356.07	2.000		9801.	9934.	9934.							
754	355.34	1.953		9872. 9922.	10004.	10075.							
131	351,40	1	2,9040	10048.	10170.	10178.							
134	3+4.4	2.024		10140.	10768.	10268.							
130	393,71	2.031		10903.	10435.	10327							
130	332.40	•••••	2.8882	10946	11062.	11062.							
134	313,40		2.7844	11842.	11975.	11755.							
137	305.30		2.8233	13417.	13331.	13331.							
139	283.24	1.995		13244.	1 1358.	13358.							
344	201.63	1.935		13334,	34429,	13424,							
344	277.34	1.976		1.8308.	1.422.	13472.							
144	274,99	1.998		13617.	13750.	13730.							
744	275.03	1.484	2 4363	13/08.	13870.	13849.							
100	207.50	8.000	.,	18071.	1 1983.	13765.							
141	165.50	1.941		14047.	14197.	18274							
1	262.00	1.921		14140.	1.307	14307.							
190	260.00	1. 910		1+2+0.	1.557.	14347.							
151	257.18	3.944		14386.	14475.	14586.							
154	234,44		2,8383		14691.	1+651.							
134	831.07	1.970		1.4447.	14770.	14770.							
199	247.75	2.01		14846.	14953.	14955.							
151	248,99	1.978		14937.	13044.	180							
150	242.86	1.996		15117.	15223.	15223.							
164	258,16	1.967		15496.	14401.	19401.							
164	235.41	2.074	3 4413	15025.	1.4474.	15514.							
16.5	248.45	2.003		19917.	14671.	15621.							
164	230.34	1,997		13664.	15766.	15766.							
163	229.54	2.010		15848.]4987.	13947.							
14/	242.20		5.6423	160.00.	16197.	16167							
144	218.34	2,000		16220.	16227.	14315.							
174	218.17	1,980		16238.	16333.	14333.							
174	214.43	1.949		14416.	16500.	16508.							
111	211.90	1.972		16922.	16615.	16615.							
174	211.40		2,8859	16545.	14634.	16636.							
175	284.82	2.000		14795.	16,882.	14882							
111	805.67	1.973		16901.	16787.	16987.							
170	202,30	2.000	2.4834	10754.	17115.	17115.							
1.00	200.54	2.043	-,	17849.	17133.	17133.							
184	178.77	2.088		17123.	17206.	17206.							
184	194.66	1.493		17813.	17194.	17396							
184	199.87	1,901		17830.	17+12.	17412.							
183	198.99		2.7824	17893.	17577.	17377.							
387	187.24	1,910		17630	17713.	17713.							
100	104.00	1.996		17767.	17050.	17650.							
189	100.00	2,028		17883.	17886.	17464.							
194	179,40	8.025		17989.	10070	18078.							
192	178,24	1,950		100+3.	14123.	14123.							
1 **	178.40	1.973	5,9040	18047.	18128.	14140.							
1,99	178.85	1		18150.	18227.	14249.							
1 96	173.30	1.971		18258.	18335.	18395.							
199	170.47	1.992		18403.	18571.	18479							
1 **	160.41	2.014			18470	14570.							
200	164.90		£,8299	.8611.	18686.	15086.							

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			3-484016	(07-39						BOREHULT	rig1-19		
	PLACE	6 - 6	DENSITY	CLANULATIVE	501 10 ##55	{'5#/C#+s21		PLACE	6 - 5	PENSITY	- UNILATIVE	SHIT MAKS	
÷	(16.07H	-Exel11	CT (0110	#PD#2.871	840*(240)	139548 581385	2	0.4 F 1 F	DENS11* (64.10)	CT SOLLO	8H0#2-823	##D#**##C>	
	5		123142			FOR PHD		•		ISH/LC1			FOR HHO
v	1800.00		2.0210				101	590.14	2.025		8496.	#43C.	8 × 1 .
*	818,51		2.9096				103		2.003			A42C,	862
	848.34		2,9191				104	585.24	2.041		4761.	#691.	8695.
1	798.10		2,9042				105	583.21	2.041		879	4853.	
	758.60		4.9257			_	10	579.13	2.019		9027		8161
1	767.62	1.950			0.	0. 18.	104	478.20 576.37	2.036	5.8343	9086.	401%. #111.	9414.
÷.	762.89	2.003		215.	209.	209.	110	513.00	2.038		9311.	4245	7245.
10	761.38	2,01*		287.	281.	241.	111	512.66	2.024		7368.	•301.	9301. 9863.
14	759.10	2.011	2,9148	392.	345.	385.	113	567.94	2.035		1545	• • • •	9577
13	757.00	2.052			485.	N#3.	11-	565.17	2.016		4725.	4660.	9660
12	733,44	2.042		687 .	675.	675.	11.	561.65	2.019			4836.	90.52
16	751.49	2.043		766.	742.	7.2.	117	560.87	2.005		**31.	996 C.	986.*.
14	749.33	2.0/2	2.9113	#85.	976		117		1.972		10114.	100.3.	100.1
19	747.17	2.291		940.	462.	962.	154	576.24	1 481		101 12 .	10071.	100-11
20	744.42	2.039		1113.	1093.	1226.	124	524.84	1,037		10220	14161.	10141.
21	138.54	2.049		1403.	1 378.	1378.	123	549.80	1.956		10.29	10374.	10274
2.5	73 . 34	2.64		1461.	1435.	1435.	120	349.00 547 90		2.7221	10468.	1-413.	10411.
5	729,80	2.55		1827.	794.	1744.	124	546.54	1.459		10:89.	1:534.	10544
24	729.90		2.4154	1045.	1412.	1812.	121	543.31	2.000		10715.	10666.	10666.
20	725.95	2		2023.	1987.	1987.	154	539.44	1.951		10846.	10851.	10841
29	124.57	2.06		2101.	2063.	2063.	150	538,90		2.7140	10929	10945.	1086*.
30 31	720.04	2.075		2275.	2237.	2231	134	334.43	1. 16 5		11105.	11065.	1104*
34	714,91	1.944		2462.	2415.	2918.	190	\$ \$3, 37	1.961		11176.	1113	11137.
33	714,15	1.992		2588.	2342.	2547.	132		1.950		11710.	11173.	11173.
35	710.22	2.042		2777.	2728.	2778.	136	528.24	1.921		11401.	11368.	11364.
50	707.00			2927.	2876.	2876.	137	526.6*	1.937		11469.	11.38.	11430.
30	702.40	2.005		\$147	3092.	3092.	134	525.92	1.745		11588.	11559.	115**
3 *	701.10	2,033		342	317.	3167.	144	522.74	1,971		11000.	:1413.	11614.
40	498,83	2.044	7,87.9	3317.	3240	3260.	144	518.41	2.003		11076	11803.	11801
+2	4 4 E U 4			3355.	3298.	3248.	1 * 3	517,64	2.621		11881.	11.57	11845.
**	492.95	3.034		3926.	3941.	3341.		515.70	1.997	2.1140	11792.	11962.	11971
*7	690.30	2.035		3737.	1673	3473.	144	513,70	2.050		12067	12049.	120.4.
	688.81 687.43	2.042		3070.	3864.	3004.		509.38	2.050		17280.	12268.	14104.
48	686.23	1.978		3*25.		30.50.	157	508.40	2.035		12500.	12287.	12287
	683.17	1.978		4047		3798.	151	104 43	2.025		12357.	12404	12404
51	679,96	8,035			+1+2	*1*2.	154	304.44	2.059		12508.	1,501.	14501.
34	677.99	2.035		4304.	+236.	4369.	134	500.34	2.000		12786.	12580.	12723.
	673.67	2.050		4520.			195	4 98 . 37	2.082		12025.	12825.	1282".
32	670,38	2.097	3 9066	4678.	4600.	4626.	15/	496.01	2.046		12942	12863.	12944.
- 57	668.75	2.104	••••••	4759.		4679.	150		2,035		1 54 36 .	1 50* 5.	1 30 4 3
	668.17	2.110		4860.	471 4 .	4719.	169	471.47	2.020		13226	13137.	13235
	463.00	2.003		\$053.	4960		263	468,35	2.030		13300.	13310.	13310.
61	661.04	2.003		5200	5059.	5459.	360		2.102		13357	1427.	12929.
	6.74.50	1	8.8401	5419.	51 52.	\$152.	164	485.01	2,113		1 3 7 7 .	1 34 72	13-92.
	657.95	2.014		5792.	3204.	5204.	147	483.00	2.847	2,7181	13255.	1,571.	13571.
	\$34,41	2.044		\$*57.	4367	5367	147	479,90	1.987		13729.	13748.	13748.
67	652.45	1.947		5548.	5+58.	5458. 1845	268	478,33 676,40	2.003		13802.	13822.	13822.
- 17	6-0.46	2.022		5642.	3551.	2951.	170	\$75.00	1.936		1.081.	14051.	14051.
70	4 4 8 . 1 4	2.034		5757. 5765	5665.	5865. 5746.	174	470,86 468,87	2.063		14142.	14162.	14162.
14	646.15	2.041		5852.	5761.	5761.	174	468,UU		2,8892	14285.	14 504	1 4 3 6 4 .
7.8	641.41	2.020		5985. 4837	5894.	3894. 5911	174	465,75	2.046		14338.	14736.	19356.
- 15	4 . 1 . 0 .	2.927		60.78	6007.	6867.	174	462.61	2.003		1.5.0.	1.557.	149-1
2.	6.59	2.04*		H173.		6083.	177	462.21	1,991		14558.	1.575.	14575,
- ֥	633.70	2.050		6.50.	6374	6374	177	439,46	. 053		1*686.	14705.	14705.
	631.00	2.011		65.5.	6470.	6478.	184	458,28	1.935		14746.	14762.	14762.
80	827.81	2.014	. 7242	84.0 84.0	6577	6377	184	133.51	2.035		14977	14996.	14946.
	677.69	2.525		6°").		6660.	184	450.81	2.022		19110.	14130.	19130.
* 1	426.31	3.014		6740.	6717. 4882.	6082.	142	146.UU	3.014	2.7201	13203.	15225.	15244
		1.972		7100.	7025.	7025.	184	446.14	1.965		15331.	19355.	1:355
	617.01	2.047		7230.	7157.	7157.	107	442,75	1.967		15473.	15500.	15500.
	613,11	2.861		7924	7352.	7352	187	437.04	2.052		15759,	19791.	1 7 91
	611.76	2.035		7461.	7811.	7911.	3.90	\$35,09	2.025		15854	15686.	15006.
94	610,80	2.050		7649.	7504.	7621.	19#	\$30.11	2.060		1.064.	16101.	16101.
98	607.25	2.003		7708.	76.39	7619.	177	428.VJ	2.053		1. 200.	16239.	14234.
- 7 <i>3</i>	604.10	1.907		7852.	7713.	7784	199	423.66	2.019		16413.	16353	14333
	682.14	2.024		79+3.	7876	7876.	190	423.Jv	2.028		16426	16467.	16467.
	578.68	2.922	2.4081	8112.	80%&.		194	419.37	2.011		16615.	16650.	16658.
	\$97.81	2,033		8150.			194	416.62	1,991		16766.	16790	16740
104	592.70	2.067		8400.	8334.	8234.					104.	180.1	1

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			BOWEHOLE	007-17						BOREHOLE	091-14		
	PLACE	5 - 6	DENSITY	CUNULATIVE		194/540051		PEACL	G - 6	DENSITY	EURULLTIVE	SOLID MASS	160/CH++7
د	UEPTH	DENSITY	OF SULID			LINEAR	J	0(51)	DENSITT IGN/CCI	DF SOLLD	##[=2.83)	Bun+ (980)	LINEAT SP THE
	DINTI	(84/22)	IGH/CC)	84085-851	WHORE WHO?	FOR RHO				(\$4/66)		*******	FOR RHU
281	415 25	1.972		16854.	36897.		501	226.90		2 708			
244	411,51	2.030		15783.	17024.	17028.	50 z	236.30		2.0541			
204	409.54	2.009		17076.	37322.	17127.	50.5	189.00		2.8975			
203	507.30	2.803	e ,///2	17169.	17213.	17215.	305	172.24		2.0387			
284	406,79	1.942		17204.	17251.	17241.	306	168.00		2,9159			
301	484.88	2.043		17293.	17348.	17360.	307	163.40		2.9133			
500	403.67	2.003		17465.	17377.	17450.	309	124.40		2,9291			
810		2.097		11483.	17524.	17528.	319	145.94		2, 73 35			
517	399.14	2.104		17564.	17508.	17388.	311	111.00		2.319/			
273	399.30		2,8403	17767.	17378.	17825.	31.3			2.8218			
41.4	393.04	2.134		7940.	17929.	17929.							
413	389.09	2.140		14058.	1.094.	18044.							
211	300,71	2.149		10226	10460.	18268.							
214	385.14	2.11#		10309.	18342.	18342.							
\$14	382.44	2,129		10.33	10464.	10464.							
221	360.50	2.151	2.9040	19556.	1	18586.							
222	380.01	2.124	-	16579,	18608.	10600.							
223	379.67	2.027		18394.	18670.	18717.							
222	375. /*	1.030		18776.	18796.	18746.							
220	372.99	1.909		18882.	18906.	18706.							
227	370.24	1.014		18797.	19014.	19082.							
224	366.43	1.649		19123.	19144.	19144.							
578	362.77	1,970		19259.	19778.	19278.							
231	341.33	1.962		14312	19332.	19415.							
23-	336.40	2.035		19549.	19943.	19563.							
234	355.50		2,4020	19596.	14609.	19609.							
233	353,73	2.046		19633	19644.	19643.							
251	3+8.42	1.976		19925.	19934.	19934.							
238	345.87	1.967		2050.	28057.	280-7.							
237	340.74	1.961		20273	20279.	20279.							
847	339.30	1.478		28.824.	20332.	20332.							
242	330,70	2.015	7,0305	2011	20922.	20422.							
	339,44	1. ***		20525.	20530.	20530.							
848	535.69	1.994		28396,	20601.	20001.							
	\$\$0.17	2.104		20768.	20773.	\$8773.							
548	320.10	2.688		20868.	20873.	20073.							
23.0	343.89	2.019		21725	21029.	21029							
251	423.47	8.016		21128.	21123.	21123.							
234	317,74	9.005		21272.	21 378	21374.							
234	316.07	2,036		21451.	21456.	21496.							
292	315.21	2,017		21307.	21513.	21913.							
231	511.49	2.053		21695.	21 02	21792.							
130	311.10		2.7973	21794.	21711.	21711.			•				
234 24U	310,50	2.057		21791.	21799.	21799							
24.3	304.70	8.050		21907.	21915.	21915.							
264	395.44	2.083		21763.	21991.	21741. 22847							
24	301.40	2.046		28171.	22188,	27180.							
263	308.47	5,905		22276	22219.	22219.							
264	299,10	3.0250		22404	22413.	22-13							
26.6	296,90		2.0733	22*16.	27424.	22424							
24.4	293.1/	2.024		224AU.	22488.	22485.							
271	290.06	2.020		22721	22787	22737.							
110	288.41	2.035		25004.	22012.	22012.							
213	283,74	1.976		27734.	22741.	2304R.							
275	282.98	1.946		23038.	23065.	23065							
276	280.60	2.017		23167.	25475.	23173.							
277	279,45	2.028		23223.	23227.	23305							
2	275 91	2.036		23345	23401.	23401.							
280	273.94	2.839		23441	23496.	23496.							
281	273.10	2,011		23725.	23533.	2335.							
283		1,925		23743.	23747.	23747							
28*	266.00	1.929		23045.	21848.	238.8.							
36.2	269,12	1,981		23733.	25783. 36037.	24027.							
20/	200.17	2,868		24121.	24124	241.24							
288	130.61	2.053		24201.	2+201.								
287	237. 9 4	5.655		24277.	26276.	24364							
291	253.94	1.970			24422.	24422.							
294	231.93	1.969		24513.	24510.	24510.							
299	250.44	1.948	1.4 70		24579.	24374							
295	248.40	1.972		246.72	24666.	29666.							
299	244.43	1,948		24740. 24848.	24753.	24841.							
227	42.50	1. 483		24938	24 930.	21930.							
194 100	243.54	2.050		24944. 29140	24483. 28120	29961. 23120.							

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			BOAEHOLL	082-64						Brat Hour	0.1		
	PEACL	6 - 6	DENSITY	CUPULATIVE	50.10 4455	154/144421					000		
5	OLPTH CIFTI	DENSIT	OF \$0110			LINCAR	J	2101-1-	CEN 111	DENSITY OF SULLO	50402171VE	TO IC MASS	- G# - C #++2
			ISH/UC)	#HOE21821	h "Gession	EUR RHA		2010	(9#222)	CONFONENT	8H1+7+821		STULNI
										1642151			FLE #40
v	1980.VU		2.8710										
1	1795.41		2.8210				307	1379.20		2.7140	7009.	7149.	1149.
;	1783.10		2,7252				100	1815 11	2.041		7227	7286.	1284
:	1708.10		2,729:				104	13/3.00	2.057		7012	7459,	7
	1778.36		8.7862				106	1310.01	2.043		7414.	7969.	7569
- 7	1760.48		8.7191				101	1364 47	2.244		7776	7758.	7748.
:	3739.00		8.7841				104	1363.10	2.19		7850.		\$91C
	1444.00		8,7340				114	1301.47	2.100		7897.	P058,	6055
1.4	1693,30		8.7100				114	1860.61	2.146		7990,	8153.	4144
14	1650.00		2,7241					1350.18	2.147		Buls.	e177.	e117,
14	1626.10		2.7140				114	1331,18	2.100		8178	#247, #345.	#24*. #345
	1616.10		2.71.40				113	2334,32	2.17		4225.		
- 17	1589.50		2.7946				117	1355.81			6272.	8**2.	
1.0	1519.84	2.194		٥.	e.	n.	110	1351.00	2.050			A644.	
20	1516,03	2.10		150.	161.	161.	120	1548.54	2.03*			8665.	264
21	1512.11	2.179		*67.	274.	274.	121	1346.00	2.045			9741. #874.	A 7 9 1 .
24	1909.34	2.144		549.	360.	960.	158	1545.10	2.003		8774	8996.	C946.
2.	1508.33	2.097		414. 700	431.	431.	124	1348.01	2 0.01		8815		****
27	1504.84			196	413.	725. 813.	123	1340.44	3.991			4181.	1101
21	1503.14	2.079		Pas.	902.	902.	121	1338.47	1.991			4742.	4247
24	1499.44	2.051		1015.	1054.	1056.	120	1357.03	1.996		9133.	9282.	4282.
24	1498,50	2.047		1121.	11.3.	1143.	154	1336,28	1.944		7173.	9364.	9364
	1473.23	2,003		1292.	1318.	7319.	134	1333.2/	2.014		\$295.	**#*.	
54	14000	••••	2,7503	1423.	1655.	1400.	134	1350,5/	2.915		9455.	4654.	
33	1488.00	2.020		1438.	1671.	167:.	134	1320.42	2.020		9966.	.759.	4749
ذذ	1485.50	2.054		1079.	1713.	1715.	135	3323.44	2.0.9		9706.	9822. 9908.	9822.
3.	1481.64	2.052		1955,	1991.	1985.	1.54	2323,70	2.0.50		9740	9994.	
	1476.06	2.000		2017.	2057.	2057	130	1514.42	1.572		9935.	19141.	10141.
3*	1474.77	2.110		2887.	2284.	2284.	137	1316.40	1.996		10130	10201.	10363
	1978.00	2.134		2897.			144	1313.32	1.992		19170.	10383.	10365.
	1468,76	2,078		2550.	2348.	2580.	144	1314,24	2.018		10230.	10404.	10404.
* 3	1467.05	2.004		2701.	2755.	2795.	143	1312.97	1.997		10291.	10501.	1.30 7
• • •	1464.90	2.044		2722.	2775.	2779.	143	1311.45	1.997		10431.	10544.	10548.
	1461.90	2.134		2796.	2059.	2854.	2.44	1309.11	2.042		10474	10387.	30389.
	3463.47	2,124		2976.	3037.	3037	1.44	1306.76	2,590		10362.	10805.	10805.
49	1428.83	2.078		3150	3152.	\$192.	144	1299.41	2.025		10669.	10494.	10044.
50	3857,38	2.122		3262.	3264.	3266.	130	1297, 75	2.003		11027.	11260.	11860
54	3454.37	2.105		32.0.	3312.	1115.	158	1271,94	2,097		11193.	11430.	11430
52	1454.46	2.180		3*60	3432.	3432.	194	1290.65	2.876		11.481.	115630.	11563.
	1451.11	2.115		1529.	3599.	1599,	133	1208.80	2,003		11+5+.	11497.	11697
54	2009.02	2.115		3693.	3644.	3644.	1 54	1204 de	2.776		11320.	11765.	11763.
21	3446.82	2.100		3796.	3471.	3471.	150	1279 44	2.140		11977	17235.	12233
	1443.81	2.434		3040.	3916.	3916.	139	1277.35	2.068		12021.	12277.	12277.
60	1441.75	2.057			4098.	4942.	161	1276.49	5.029		12149.	17987.	12404
	1+34.29	2.049		NU96.	\$177.	4177,	368	1211.11	2.278		12299.	12521.	12521.
6.3	1457.00	2.037			4263,	4267.	164	1270,05	2.278		12508.	12775.	12670.
*5	1943.44	2.011		4348.	1434.		363	1269.34	2.273		12559.	12828	12424
64	1441.85	2.047		4492.	4517.	4317.	144	1264,84	1.974		12037.	12881,	12081,
	1428.42	2.030		458.	4750.	4750.	140	1261.04	1.930		12914.	13191.	13191
5.4	1424.56	2.0**		*742. ****	4656.	**56 .	167	1230.00	1.917		12978.	15246.	132.8.
70	1423.10	2.015		***7	4983,	4745	174	1257,10	1.897		232.88	1 44 5 7 .	13917
14	1470.47	2.040		494 8 .	3046.	5846.	174	1834.64	1.894		13170.	1 34 53 .	13453
7.5	1418,90	2.01*		5115	5818	5812.	174	1293.74	1.917		13278.	14520.	13326.
15	1418.33	2.020		5231.	5255.	5233.	178	1251.00	1,911		18851.	13459.	13631.
76	1414.67	2 009		\$414	\$975. 8819.	3274.	176	1290./*	1.946		13908.	1 1658.	13658.
7.	1412.91	2.018		5395.	5507.	5562.	170	1298.16	1,741		13465.	13755.	13755
77	1410.42	2.040		5497	3564.	5344.	174	1246.94	1.927		13591.	1 3012.	18415.
μu 	1404.54	2.045		5394.	5667,	3467	180	1242.98	1.946		13744	13907. 160 6 1.	13907.
	1475.6/	2.094		5485.	3797.	\$797.	184	1257.43	1.741		13896.	1.196.	19196.
4.5	1474.39	2.0		5011.	3862,	3862.	18.8	1238.57	2.100		13992	14294.	14245.
	1404,57	2.018		2011	5926.	3726.	.83	1229.24	2.113		1 . 5 . 5 .	14531.	14531.
	1399.49	2.005		5974.	4992.	6892.		1229.10	£ . 1 . 1	2.7140	1-404	. 721.	14721.
14	1397.20		2.7262	6151.	6273.	6273.	387	1227.13	2.129		14526.	4732.	1 • 7 • 2 . 1 • 8 • 1
	1395.21	2.052		6156.	6278.	6278.	187	1217.69	2.076		1 750.	\$070.	1.070
9 0	1343.43	2.047		6344.	6363.	6363.	190	1237.48	2. 425		15262	5380.	15346,
71	1392.07			6386.	4513.	4513.	192 1	1210.02	2.167		15440.	3827.	15827.
*>	1389.17	1.477		6489. 6328	6618.	4418.	190	203.55	2.132		15001. 1	6166.	16144.
••	1388.51	1.967		4368.	6677.	4437. 6477.	194 1	1200.95	2.115		14010.	6240. J	6379.
94	1308,79	1,980		4446.	4778.	6778	195 1	197.09	2,003		14119. i	6167.	
91	1384.44	1.999		6743.	6810, 4879-	6818. 6879	197 1	191,94	2.090		1-227. 1	6582, 1	6582.
78 77	1 588.79	2,087		6786	6985.	\$721.	170 1	187,40 182.4/		2,7242	14626	6997. 1	6787
: 70	388.78	1.017		4731.	7049.	7869		175.85	2.926		17002. 1	7371. 1	7971.

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			BUREMOLE	087-04						BOREHOLE	082-84		
	PLACE	6 - 6	JENSLIT	CUNULATIVE	SO. 10 -455	16#/2=+21		PLACE	6 - 6	UENSATT	CUMULATIVE	50110 =455	
J	DIFTI	DENSITE 164/CC+	CP SULID CD#PONLNT LJM/CCI	9HQ82+82)	RIOUCRMO>	LINCAP BPLINC For RHD	J	D1 # 1 #	NEWSII (G=/CC)	0F SOLID C34PONENT (64/CC)	#HD#2+\$21	RHUTCHMO>	11N(4) 5P(1N) FOR RHC
4 11 -				17864	18251 .	18251.	38 1	180,/8	2.170		38473.	54923.	38921
20 X	1341.47	2.2		10375.	18773.	10775.	50.	779.74	2.138		58526.	38965	38964.
204	1106.32	2.174		18733.	19877.	19344.	505 504	765,/6	2.175		39267	39717.	4971'.
105	1145.10	2,397		19421.	14740.	19760.	305	765.61	1.750		59350.	34800.	3780C,
20.	1141.40	2.403		19719.	20147.	20147.	306 30 (737,14	2.107		39414.		
	11	2.100		19983.	20417.	20417.		734.60	2.1*3		39750.	****	
267	1127 36	5.002		20363.	20606.	20805.	310	744,39	2.147			40512.	.0.12.
21.4	1128.4	2.033		20576.	21024.	21024.	311	743.44	2.047		+0#92. +03#3	NO772.	
574	1121.00		2.8210	2064	21107.	21107.	213	7 59, 36	2.00#			40753.	
414	1120.14	2.054		20701.	21151.	21151.	314	757.42	3,970		40602. 10781.	*1072. *1 23 1.	41231.
215	1113.44	2.035		21060.	21510.	21710.	316	**1.**	2.076			41386.	1.1.1.1
217	1110.44	2.078		21209.	21725.	21725,	314	744.99	2.073		+1202.	41672	
21.9	1145.14	2.1/*		\$1247	29041.	22841.	31 4	741.90	2.091		41894.	*140*.	*184*
424	1102.67	2.105		21413.	22043.	27210.	320	736.40	1.996		+1+17.	42167.	-204
224	1098.38	2.07.		21834.	27244.	22285.	372	714,44	3.953		41794.	*216*.	****
224	1041.44	2.007		27156.	22608,	22408.	\$25	710.84	4.01 3		41878.	42527.	42527
287	1083.7-	2.059		22460.	22910.	22430. 23258.	325	789.13 786.96	2.011		41761. 82066.	42411.	+2514
224	1044.17	2.030		73452.	23902.	23902.	\$21	785.01	2.071		42128.	*2578,	42518.
224	1062.16	2.0**		23418.	24868.	24262.	524	102.2	2.177		42457.	\$2904.	
	1057.10	2,192		28905.	2+395.	24345.	330		2.030		\$2722.	*3172.	*3172.
431	1053,31	2.120		24118, 24226.	245576.	24676	331 534	641.01	1.900		±3002.		
233	1047.75	2.037		24369.	24637.	24839.	555	643.35	3.685		45185.	*3635.	43435.
234	1045.41	2.104		24833.	25338.	25838.	335	677,55	2.036		43394.	*38**	
236	1034.00	2.115		25111.	25561.	25561.	330	674.71	2.004		43518. 	43968.	+396F.
234	1030,14	2.004		25349.	25997.	23944.	490	665.33	1.919		\$3965.		
237	1020.27	2.126		25016.	26266.	24266.	854	663.61 661.47	1.927			44487. 44583.	*****
241	1019,97	2,150		26042.	26442.	26442.	341	661.04	1,950		44152.	***02.	
242	1012.11	2.170		26252.	26795.	26745.	343	634.1/	1.9**				
244	1004.10	2.1+1			27025.	27025.	344	631.1/	2,003		44397.	45047.	*****
244	1000.07	2.134		26890.	27340.	27340	344	641.50	2.241			*5528.	
247	997.54	2.186		27027.	27677	27477.	34/	637.00 648 BU	2.004		45299.	45749.	45847.
343	987.24	2.075		27548.	27998.	27998.	347	433.5/	2.124		43463,	45915.	42915.
834	983,74	2.193		27729.	28179.	24412	190	601.45	2.130		45740.	46076.	46190.
252	972.68	2.232		78857.	28802.	28407.	\$54	625.94	2.990		4.447	46347.	-6347.
234	967,71 969,78	2.104		28922.	29238	29234.	354	623.27	2.117			46632.	46612.
233	763.84			28995.	29445.	29465.	337	618.77	1.122		46241.	46691.	46641.
234	938,44	2.86*		27534.	29984.	79984	351	613.4U	2.056			****2.	44927
120	**7./3	2,103		29665.	30115. 30313.	A0313.	358	609.74	1.953		46769.	47103.	**219
264	930.14	2.129		30576.	31020	\$1928. \$1881	560	605.25	2.073		16893.	47302.	47302.
261	980.21	2,159		\$1134.	31584	\$1984	344	578.38	2.201			. 767.	.7675
263	918.12	2.1*3		31450.	31700.	31700.	563	593.43	2,195		47588. 47577	47958. HA026.	47958.
245	915.14	2.335		31406.	31856.	\$2856.	360	587.65	2.073		.7795.	44245.	
266	918.24	2.187		\$1540. 31400.	3;990.	32038.	364	505.94	2.245		47986.	48336.	
260	908.25	2.181		11766.	37216.	32216.	368	\$74.11	2.141			48954	*895*.
210	894.94	2.129		32447	3,947.	3294 1	367	567,05	2.167		48927.	49377	
271	877.34	2.036		32519.	52968.	32968. 85271.	371	561.70	1.970		49384. 89262.	*****	*****
	\$\$2.70	2.155		55103.	33953.	33993.	373	559.75	2.047		49283.		44735
274	801,21	2.1*1		33395.	33806.	33806.	379	554.13	2,107			50020.	50020.
274	875.44	2.141		33516.	33968.	3396C.	376	\$32.56	2,130		49462.	50112.	50112.
274	878.71	2.115		33748.	34192.	34142.	578	544,/4	2.871		50078	50526.	30526.
279	464.54	2.144		33679.	34329.	34329. 34443.	579	542.30	1.845		50174.	58624.	50624.
587	844,47	2,150		14065.	\$1.335.	\$\$535.		538.27	1.625		50 143.	50793.	50793.
282	868.19	5,135		34914. 34446.	34764. 34896.	39896.	582 583	536.13	2.064		50430.	50888. 51661.	51961.
284	433./*				35086.	35086.	384	521,90	2.093		51160.	1610.	51418.
285	851,6V 549,07	2,098		34744. 34876.	35326.	35326.	983 384	515.30	2.146		51782. 51496.	51946.	51946.
e# /	44 * , 21	2,950		54762.	34412,	55412.	381	509,54	2.098		51812. 	52262.	52261.
289	841.75	2.183		15230.	35488.	35688.	387	503,76	2.140		52167.	\$2557	52557.
	834.15	2.148		33535.	35985.	35785. 36750.	590	497.95	2.255		52443.	52893.	52893.
274	827.4*	2.047		35883.	36333.	\$6533.	190		2.001		\$2774.	53224	33224.
293	827.94	2.036		39348,	36398. 36614.	36514.	 	489.3/ 486.79	2.074		52705. 55436.	53586.	53555. 5 3466 .
291	818.33	8.150		36*31	36-01.	36891.	395	485,54	2.115		53102.	59552.	53552.
294	815,12	2.127 2.107		34915. 36938,	373087.	37388.	391	475,20	2.167		53638.	54087.	94087.
274	501.25	2.150		37247.	37597.	37547. 17848	390	471.11	2.078		53018.	51268.	5426A.
10	793,31	2.12		37687.	36139.	38119		466.74	2.075		SHORD.	\$4530.	54550

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			BORF HOLE							BOREHOLE	001.00		
				CUMULATIVE	10 10 MASS			PLACE	• • •	DENSITY	COMPERTING	20110 H435	147/24421
J	D1	DENSIF	CF SULID. COMPONENT	H40#2+871	-	125648 571895 For 840	L	2001 2000	(68)124 (68)124	COMPONENT ISH/CLI	#HU#2.871	#1-#<##03	SPLINT FER AND
			(G4/LC)										
401		2,094		44211. 88473	54641. 54822.	54661. 54P22.	u 4	1800,00	2,12!	2,8210	۰.	٥.	۰.
4C∡ 403	453.0V	2,203		54492.	5. 942.	54547.	•	996.10	2.16		65.	63.	67.
	456,75	2.147		5*611.	99961. 99961.	55244.	:	992.4	2.159		276.	276.	276.
403	471.1	2.1**		54935.	95367.	15161.	•	988,7U	2.073		443.	463.	463.
497		2.158		5587'.	54571.	49978	,	182.11	2.175		776.	* * *	
400	434,34	8,813		55647.	56147.	56147.			2.170		1015.	1015.	1015.
414		2.828		55841	56291.	16 17 3.	19	972.50	2.041		1322.	132.	1322.
•11	\$26,70	2,102		54157.	34607.	36677.	14	369.41	2.159		1*6?	1.75	1776.
	423.27	8.107		56334.	55309.	57109.		960 12	2.212		1 465.	198°.	1961.
*12	415.97	2.035		56701.	\$7151.	57151.	1.	939.11	2.260		2019.	21	23.1
	\$13.04	2.037		56805.	51 551.	57537	1.	936.44	2.19:		2235.	2	22.4
	\$07.39	2.018		57132.	57962.	51562,	10	973.21	2.308		2919.	2-19	
414	405.45 807.8/	2.132		57345.	51195.		17	931.33	2.335		2541.		2561.
421	346.01	2.177		57554.	58004.	2800%. 98074.	27	947.16	2.:22		2757.		
424	397,22	2.105		57836	38286.	54286.	24	944.81	2.156		2932.	5932.	2942.
			5.8510				23	936.34	2.122		33.4.		
							22	932.77	2,124		5561.	3771	1771.
								927.28	2.175		5857.	3857	3847.
							54	926.10	2.131		3921. 4111.	3721.	4:11.
							30	920.34	2.232		+e21.	*223.	
							31	918,62 816.07	2.279		4940,	4540.	
							3.5	412.14	2.144		.680.	4690.	9490. 6890.
							54	409,90 906,81	2.1**		5905.	A005.	5005.
							58	902.01	5.148		5274.		2224.
							37	876.77	2.180		5551.	÷**1.	5531.
							37		2.151		5680.	*6e".	5020.
									2.105				
							**	856.34	2.111		6127.	6127	6294.
									2.277		** 51.	4452.	4451.
								878.40	8.219		6748.	6743.	6783.
								\$75.55	2.206		6853.	6853.	6851.
								870.20	2.213		7432.	79.32	7032
							30	868.23	2.222		7143.	1143.	7143.
							34	864.27	2.255		7374	7374	7 5 7 4 .
								868,54	2.730		7647.	7647.	7647
								\$35.63	2.234		1075.	7875.	7814,
								830,14	2.2/6		8205.	R205.	
							51	548,74	2.247		827%. 850s	A275.	8274.
								843.4¥	2.260		8573.		
							61	447.44	2.227		8802.	8667. 8802.	##n2.
							6	839,UV	2.24		8847.	RP#7.	8867.
							63	\$15,1b	5.510			9174	9014
								832,01	2.197		9252.	#252. #318.	9252.
								\$29,25	2.535		9412.		9912.
								825,/1 824,74	2.335			4635.	1644
								\$23.33	2.257		9778. 9895	9778. 9878	7774, 9824.
								828.24	2.960		****		
								6 817,44 6 815,44	2.205		10296.	10123.	10256.
								813.70	2.178		10122.	10322.	10322.
							7	/ 811.93 D 809.1/	2,100		10984.	10544.	10544.
							,		8.100		10779.	10779.	10774.
								1 800.31	2.175		11043.	11063.	11063.
								2 798,94 8 797,36	2.264		11153. 11246.	11153.	11193.
							:	795.07	2.267		11462.	11342.	11362.
							:	5 792,69 6 791,04	2,195		11522.	11612.	11=22.
								1 749.64	2.210		11479.	11679.	11679.
							:	9 786,34 9 784,76	2.22/		11973.	11002.	11973.
							2	0 782.81	2.230		12134	12134.	12134.
							;	2 778.07	2.332		12369.	12369.	12367.
								8 776.50 4 774.92	2.143		12968.	12569.	12969,
							÷	. 772.95	2.3**		12693.	12693.	12695.
								6 778.59 7 769.91	2,834		12930.	12930.	12930.
								8 766,26	1.127		18091.	13091.	13041.
							30	* 761,74	2,205		18548.	19390.	1334A.

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			e jar en j	at 1+05						BARCHOLD	0.1+05		
	PLACE	6 - 6	0145177	CUMULATIVE	SHLIC HASS	(6=/(=++2)		PEACL	6 - 5	DENSI1Y	Company attac	SO ID HASS	(58/683)
•	01011	160/001	COMPONENT	AH0=5-851	-	SPLINE	J	D15 T1	DE45111 (64/CC)	OF SOLIC	#HD=2+821	Rubection	LINEAP SP-1WF
			184/601			FOR RHO				154/001			FOR PHC
101	739.76	2.273		13466.	1	13466.	401	328,00	2.206		27210.	37316	****
102	750.80	2.213		13765.	13650.	13765.	202 204	528.07	2.238			27255.	21255.
104	732.48	2,170		18497.	13897,	13847.	20-	524,75	2.216			27393.	27446
106	747.15	2.254		14163.	3 - 143.	14163.	206	521.7/	2.255		27575.	27414.	27513.
	743.46	2.219		1 ** 7 8	1	14473	500	519.41	2.291		27761.	27761.	2.7.1.
110	7 30 70	2,241		1-544.	14494.	14544	204	516.24	2.225			27948.	279
314	735.10 730.03	2,222		14904.	14904.	14964.	214	511.*3	2.301		24207	28135. 28207,	20207
110	729.23	2,202		15243.	1 - 2 - 3	15245.	212	511.14	2. 12		28255.	28255.	28244
115	725.81	2.291		13471	14471	15471.	519	507.60 506.02	2.250		28471.	28471,	28477
117	721.17	2.291		13591.	15571.	15541.	216	50	2.272		20.53	24635,	48633
314	717,01	2.293		19455.	14855.	15845.	-1	501.30	2.264		20046.	28751.	28741.
120	715,87	2.241		16047	16047	16047.	220	497,36 493,03	2.265		29079. 29326.	24074.	29074.
122	709.96	2.772		1+399.	16100.	16377.	223	491.05	2.202		29435.	24434.	294 15.
15+	705.64	2.315		16785.	16542.	16947.	225	487.52	2.301		29645.	24573.	29645.
125	702.07	2,507		16882.	1448.	168#2.	227	482.74			29788.	29788. 29931.	217AR.
121	695.00	2. 132		17314.	1 7 1 1 6	17316.	220	480,U4 479,23	2.271		50047. 80388.	30097.	300
1.24		8.301		175 11	17535	17933.	229	*78.**	2.301		\$0191.	30191.	30191
734	689,88 686,34	2.205 2.301		17628.	17628.	17628.	2 5 4		2. • 56		30+97.	3n365. 3c+47.	30364
134	685,37 68, 91	2,510		17692.		17842.	234		2.424		30630. 30761.	30761.	30630
13.	6.00.44	2.501		18209	14209.	18/0 .	235	468.62	2.411		308+C.	308+0.	
130	676.30	2,310		10.51.	1=330. 18+51.	18330. 18451.	235	463.11	4.491		31208.	31208	51206.
130	\$74,75 670,98	2.393		18798.	1857%.	38574.	231	459.17	2.368		319362.	31336.	31336.
139	670,39	2.32'		18823.	18823.	16823.	2 S B	*37.94	2.356		31536.	11530.	51538.
1 4 1	445.00	2.224		1 1 56	14156.	17156.	240	455.21	2.337		31840.	31840.	31840
143	637,60 636,0e	2.225		19386.	14586.	195 8 5. 19676.	242		2.250		37147.	33 484. 32169.	319R4, 32369.
143	625.2/	2.294		19#38.	19838.	19818.		445.37	2.246		32262.	32262.	32262
146	648.54	2.246		20125	20124.	2012*.	245	441.46	2.244		32559.	32559	\$2534
1.44	6 . 5 . 82	2.123		20371	2 . 3	20254.	24.1	\$ \$5. 55	2.2		32493.	32873.	82728. 52893.
349 334	643,83 442,74	2.341		20418.	20505.	20418. 20585			2.348		33107. 33249.	33102. 331*9.	33102.
191	639,88 637,91	2.254		20549	20499	20549.	201	475.51	2.285		335j6. 	31116	\$\$516
154	435.98	2.132		289-6	21446	20944	252	*15.**	2.075		14030.	34050.	34050
155	629,45	2,238		2143°.	21210.	21037. 21218.	25		2.112		34398.	54276. 34348.	34274.
199	620.90	2,268		21337	21557.	21537.	235	404.04	2.081		34499.	164 ···	34444
154	619,91	2.191		F1F11.	21811.	21811.	231	482.09	2.086			34741.	3474 : .
160	616.65	2.254		22746.	21946.	21946.	255	396.51	1.000		34 979.	34 474	34474.
144	611.00	2.825		22460.	22044	22084.	261	391.06	2.03		33260.	15116. 15267.	35076. 15267.
100	607.99	2.272		28476.	22476.	32474.	24.4		2.613		55661.	****1.	33641.
1.6.8	68¥,48	2,124			27809.	22804.	263	378.07	2.065		35888.	34.848	35444
167	\$99.44	2.341		28403	23005.	22005	269	370, VH	2.035		34233,	36233.	34215
147	595,74	2.321		23227	23104.	2822	264	369.41	2.035		36316.	36290.	3631C.
171	591,89	8.343		23301.	23301.	23301. 23474.		363.90	2.1.5		36943. 36982,	16443. 36502.	36983.
174	590.2/ 587.93	2,400 2,349		28374.	23374.	23516,	87x	858,44	2.149 2.068		36729. 36871.	36729.	36729
174	584.73	2.348		23072	2	23802,	273	855.63 350.9u	2.076		37869.	37009.	370.4
1.15	581.48	3.374		24134	24114	2*13*.	213	348.94	2.106		37458.	17358.	37358.
174	574,98	2.501		24328.	24395. 24524.	24355.	271	845.UV	2.587		37566.	37566.	5746; . 5756f .
179	572.10	8.270 8.301		24696.	24696.	24696.	279	343.US 341.46	2,112		37668, 17149.	37668.	37668.
114	587.55	2.229			24978.	2.978.	280	339.00	2,120		37831.	3-831	370.11
184	564.87	2.507		25142.	23042.	20042.	284	336.15	2.11		37996.	37994	37851.
187	568,40	2.297		23283. 25404.	23283.	23283.	284	343.17	5.145		38037. 38182.	3803'. 38182.	30017.
386	557,44	2.235		25545.		25545.	582 582	331.44	2.018		38284.	38284	38244,
	353.66	2.224			2	25746.	28/	326.07	2.02+		38.96.	38+96.	38+96.
190	549.78	2.178		25733. 26920.	24933. 26020.	25933. 26020.	284	321.77	2.144		38616. 38763.	38614. 38763.	38616. 38763.
145	544,87 544,81	2.199		26195,	26145.	26195.	547	317,01	2.150 2.046		38805. 38907.	34805.	34405.
195	543.88	2.227		26341	26343.	26391	292	317.44	2,034		38943.	14483.	38967
1 9 3	558.70	2.21		26540.	26706. 26640.	24448.	294	314,64	2.067		39119.	39119.	390%). 39119.
197	537.71	2.206		26684. 26748.	26684. 26798.	26684. 26798.	-	311.14	2.1**		57236. 39299.	34238. 34244.	37238. 39299.
198	552.75	2.212			26868	26844	291	308,37 307,24	2.097		39944	39464	39444
200	\$30.*3	2.208		27121.	27121.	27121	244 30u	304,47 303,66	2.061		37646	34646.	39646.
											37685.	3468*.	39685.

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			BOREMOLE	067-05						SOAL HOLE	048-74		
	PLACI	6 - 6	DENSITY	CUMULATIVE	50:10 MASS	1681280025		PLACE	6 - 6	01-5177	COMOLATIAN		
-	12 F 1 F 1 F 1	26 NS 11	OF SOLID			L 24/5 ##	÷	OFFIN	etvetie	0 50,10	C001 811.45	20110 4=22	LINCAR
			164/101			FOR BWD		0.64	11, سير ا	COMPONENT	AMU12. #21	RHORCENC,	SFLINE
													FOB BHE
391	100.91	2.0.		39840.	14840.								
304		1		19932.	***32.	39937.		993.51		2.8210			
304		1.91-			60079,			993.3D		2,7303			
101		1 11-				482DA.		984,UU 864 VU		2,1201			
347					- 3302.	40307.	,	\$44.00		2.7483			
338	44,13						•			2.1762			
101		2.868						433.50		7,7381			
111	298.17	2.83		45794.	405 94 .	48344,	7	\$24.20		2.8516			
334	208,84	2.065		.0790.	49798.	.07.0	1.4	#19,40		2.0950			
314	279.23	2.067				SUBAR.	14	875.40		6.6631			
313	273.74	2.091		01170	4 1071.	41071.	13			2.9834			
316	272.16	2.83.		+1205,	.1205.	4120*.		844.90		2.8973			
310	264.02	2,171		•1366,		41366.	i+	855.40		2,9142			
31.9	766.63	2.171		.1.93	*1993.	*1**3.				5 . 41 . 3			
324	265.34	2.070		41660,	.,	4164F.		\$24.3U		2.4:47			
322 322	129.21	2.040		41680, 	41680.	41680.	50	811.00		2.9196			
325	138.37	2.0		41712.	41412.	41917.	21	801.50		2,9237			
354	237.24	1,947		41766	N1968.	\$176E.	23	783.40		2.9139			
328	252.44	1.91		42058.	×2058.	4205P.	1	774.40		2.9749			
521	248,34	1.772			\$2325.	42325	25	764.80		2.9253			
324	246,37	1.447		42898,	*2.98.	.2348.	21	744,40		2.9183			
530	243.84	2.112		+2322.	+2522.	92982.	24	734.30		2.9255			
354	242.44	2.215		.25.86	+2344.	42586.	30	719.90		2.9253			
333	241.09	2,260		42455.	*2455.	42645.	5.	706.34		2.8882			
, ,.	237,52	2.388			*****	2880.	34	694.60		2.9113			
212	753.35	2.508		42986.		42986.	33	675.14		2.8974			
	251.01	2.042			41087.	430#7,	32	+53.42		7.0584			
۰. ۰	156.04	2.123		3268	*3268,	45268.	50	643.60	•	2.9090			
• • •	287.00	2.144		• 3= 1 1 .	*3*12,		30	\$35.50		2,90+6			
344	211.11	3.414		435475	43547.	4 5 5 6 7 .	37	\$22.74		2.9199			
554	219.00	2.005		43766	* 3 * 6 6 .	43766.		630,94 609,5v		2,9043			
344	211.00	2,109		*3956.	43858,	43858.		540,50		2.8478			
548	211.94	2,746			44164,	44364	و ب	501,20		2.8639			
3 % % 4 % 7	209.17	8.142		***1*.	44318.	44318.		566,34		2.7344			
	202.*'	2,315			*****	4+390, #+#95.	**	564,86		8.7281			
347	282.09	2.124		44697	44697.	44647.		554,00		2.7140			
3.1	177.00	2.1.57		44 9 02.	*****	44862.	47	530.40		2.7633			
1.1	1 97.40	2.100			Na 451.	****1	50	517.64		2.7739			
154	193.34	8.134			N9017.	45046,	54	517.00		2.7932			
\$3.5	191.04	2.141			N9383. N9286	43184,	14	512.30		2.8108			
	196.47	2.11*		+5327.	49.527	45.827		302.1-		2.7845			
550	142.44	1.952			48432.	****	56			2.7343			
154		-	8.8810		••••			200,50 578.70		2.7148			
							59			2.0434			
							60	436,38	2.101		٥.	٥.	٥.
								*33.0/	1.680		56.	54.	56.
							6.5	\$91.91	1.959		149.	190.	140.
								449,27	2.022	1	297.	297.	294.
							6.6	**7.4*	2.149		115.	350.	358.
							67	445, <i>8</i> 0	2.117		\$19.	323.	323.
							47	441.42	2,203		730	71 .	719.
							24	437,10		2.7447	455.	867.	
							14	41 53	2,022		\$73.	887.	887,
							7.8	435.20	2.0%6		1043.	1060.	1945.
							7 •	434,\$0		2.7188	1988.	1107.	1107.
							76		2.030		124.	1163.	1147.
								4 50 . 70		2.7323	1266.	1264.	1268.
								427.70	2.215	2,7467	3437,		13.0.
							RU.	426.26	2.071		111	1373.	1392.
							#1 8.4	424./1	2.106		1342.	1619.	1614.
							8.5	.20.04	e.147	2 76.50	1736.	1763.	1763.
								420.05	2.106		1436.	1444	1466.
							84	419.00		2,7385	1942.		1974.
							• /	416.94	2,117		1798.	1440.	1990.
								425,84 418,67	2.085		2078.	2112.	2112.
							10	411.07	2.070		2213.	\$250.	2250.
							94	410.64		2.8016	2417	2287.	2313.
							72	488.74	5.048		2370,	2.04.	2406.
							94	487,23	2.097	2.4043	2440.	2475.	2= 75.
							73	NO6.06	2.051		2348.	2562.	2542.
								488.42	2,030		2565,	24.96 .	2476.
							78	\$97.51	2.658		8949	2995.	2995.
							144	372,87	1,913		8897,	3882.	3842.
										*. ****		3182.	#1+2.

			81414016	08H-7A						BUREMOLE	DAI -24		
J	PLACE SEPTH DIFTI	6 - 6 DLNSJ1+ (6m/CT)	154/651 154/651 154/651	84085*851 COMOFF11AC	50:10 #455 Auguranos	(67/00002) L14088 5PL340 F08 440	J	PLACE DEPTH DIFT	6 - 6 DENSITT 164/LC1	DEWSIIT DE SOLIE CGEPONENT IGN/CCI	EU ^{HU} LATIVE H ^M D#2,821	€0:10 #455 €но±€рм0>	164/(****) Linte Stint for enn
101	393.24	5.301		3167.	3190.	3190,	201						
104	372 10		C,8299	3226.	3754	32.00	204	235,11	2.0.7		108.5	10870. 10839.	10880.
104	590.32	2,116		3308	3131.	3331.	503	232.44	2,051		10892	10896.	16896.
100	588,58	2.073	5.44.4	3412.	****	3440.	205	291.41	1.997		1100-	11904	11009,
101	587.02	2.078				3509.		225.30	•	2.8164	11350	11144.	11324
109	383.34	2.036		5663.	3681.	2601.	504	225.01	2.017				11548.
310	381.47	2.043		37.8.		2795. 38w7.	510	219.96	2.035		11.44	11389.	11549.
114	178			••••		3920.	214	\$15.20	2,046		11747	11760.	11 %* .
119	376.53	2,022		1498.	4V12.	·····	213	211.50		2,0476	1799	12002	1. 00.
112	373.00		2.8768	*129.	+1+1	*1*1.	215	209,00	2.019		15386	12062.	12142.
317	372,48	2,152		4252.	• 262.	4262.	215	206./5	2,030		12220	17250.	122.47
118	568.76 866 00	2.009		*3**.			514	202.66	2.006		15014	1241	1230-
120	363.71	2.0.12				***5.	220	201.20	2.006	7.7613	12492.	1	12446.
121	361.30 361.20	5.039	2.7866	4771. 4780.	. 781.	*781. *790.	221	198.20	2.038		7635	174	12645
150	360.00		2,4545	*# 59 .		4850.	223	196.25	1.927		12726.	12641.	1.647.
122	377,40	2.005	2.8700	4778.			224	193,54	2,022		1244.6	1 7 8 7	126*
126	357.11	2.022		49#2. 5814	•••1. •0•7	4941. 3047.	225	189.65	2.003		130.43	1	13034.
154	333.50		2.7760	5460.	5069	5069.	228	187.34	1.964		19047.	13074.	130
130	335,16	2.093		5145.	5086.	5202.	229 254	186,35 186,3V	1.924		191 . 3	1 1180.	1314
131	352.04	2.046		5234.		5266.	234	106.24		2.8496	15187	13190.	31341.
134	348,17	2,014		5+25.	54 57	5 \$ \$ 7 .	232	284,99	1,793		15/33.	13291.	152-1.
130	346,60		8.9249	5502.	*513.	5513.	234	101.11	1.880		13361	13387	133915
1.36	330.07	1,903		3504.	5913.	5913.	2.38	174.00	2.005		13430.	33436. 14421	13496.
157	337.00 336.32	1.972	2. 10. 3	3916. 5972.		5724.	237	176,83 178 34	2.915		13570	13575.	13.7.
1.54	334.10	1,916		6875	6082	4882.	234	175.90	2.034	2.8255	13706	13682.	13682.
1.44	332.64	1,927	2.7066	6141.	6130.	6150.	240	172.94	2.059	2 8498	13134	13739.	137.4
145	331,85	2.001		6176.	£185.	6145.	247	172.60		2.8791	13770	14774	13761
1	225.03	2.759			4446.	6446.	2.44	171.01	2.021		11741	11196	13746.
145	323.00	2.547	2. NA 92	6 4 A 7 . 6 4 7 6 .	4495. 6502.	6495. 6502.	24.5	169.95	1.967		13918	13921.	13921.
1	372.1*	2.0%		6629.	6634.	66 34 .	241	164.75	1.42		19787.	13484.	13989.
1	317.09	2.070		6477	477	6879	244	162.46	2.090		1 223	1 = 2 2 2 .	142:7.
150	319,44		8.8410	6763. 6796.	4964. 1997	6964. 6947.	250	197.00	1.975		1	14443.	14245.
154	313,99	2.045		70	7036	70-6	254	155.41	2,045		1452C.	1-515.	14515.
194	310.10	2.670		7220.	1228.	7220.	252	232.2# 130.4V		2.9690	1 * 1 1 7	14700	1 . 708
150	304,15	2.020		7266.	1265.	7265.	233	1.4.9 . 6.4	1.644	******		14878.	14778.
15/	307.50	2,003		1559	****	1347	251	1	1.947		1**63.	1	14950
199	305.00	* *	2.8768	7+35.	74 32	7+52.	254	145.10		2.4997	1 * 26	1=112.	15012.
191	309,60 104.66	2.080	2,8146	7444.		7441.	264	1+1.87	1.948		13+68	1*038.	150 .
164	\$01.00	1.474		7	- 5 3 5	76	262	138.40		2.8498			
164	298.60	5.144	2.8508	7786	1782	7762.	26.3	135.90		2,8948			
365	276.87	2.111		7877.	7872.	1872.	265	130.00		2,9275			
367	294.00		2.7803		P020.	8020.	266	120,20		2.9341			
169	273.00	1.967	2.8498	8070, 8141.	8766.	8066. A117.	268	110.69		2,8710			
170	298.67	2.114	••	4183.	.174	41 **	267	.05		5.8510			
174	287,95	2.097		8323.	8239.	4314							
173	287,50	2.085	5.0341	8346.	6341.	8341.							
175	282.94	2.036		8379	457.	85-6,							
177	279,79	2.017		871C. 8728.	870%. 8723	8704.							
170	275,91	2.100		892E.	8922.	8922.							
3.80	78.47	2.006		7214.	4278.	9208.							
181	268,52 267 BU	1.995	2.81/1	9305. 9319	9298.	1248.							
183	266.97	2.062		9380	• 3 * 3	9373.							
182	262.10	1.983 2.870			\$*\$'. \$579	9467. 5574.							
186	260.57	2,006		94	+493	9643.							
1.8.8	294.01	2.015		• • •	4768	9768,							
190	236,89	1.959		9896. 9968		9846, 974							
191	282.90	1. 159		1003-	10136	100 14							
142	252,57	1.975		100-1	10094	10190							
194	850.46	1.896		10155.	10158.	17140.							
194	1=8.47	1.95*		10451.	10242.	10.36							
190	241.34 240.5V	1.927	2.8343	10537.	10541	10541.							
199	239.19	1.965	• • • • • • •	10672	1-527	10627							
	430.01	1.000		100241	19663.	SUNKS.							

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			8040 HOL (017-11						BORTHO, C			
	PLALL	6 - 6	DENSITY	CUMULATIVE		167/284421		BT AV S					
	CIPT.	CENSIII (60/CC)	DE SOLID	#HL #2. 821		LINCAR	ر	DEPTH	CENSIT	DENSITY DE SULIL	CUMULATIVE	\$0,10 ##\$K	1007004021
			168/101		*******	FUR RHC		D-+++-	164/021	CONPONENT	BHC=1.821	BHDRCRM'>	54 140
													FOR HHP
	1000,00	2.851	2.8210				101	252	2.011		9269	- 26 \$	
	×24,U9	2.071		39.			104	236.57	1.950		9875	• 173	3177
:	*32.71	2.014			98.		10-	224.44	2.057		****	4448.	
•	419,47	2.074		275	814. 375	# 5 4 . 2 7 5	100	212.35	8.041			4767	141
•	417.04	2.049		Aug.	392.	392	196	220.10	2.038				7861
	423.88	2.002		484.		489.	100	216.44	2.044		19056	9979.	9975
•	411.90	2.094		662.	442.	462.	109	215.95	2.036		10045	10095.	100.
11		2.002		736.	736.	236.	111	211.01	1.992		10142.	10152.	1015.
14	405.90	2.083		727.	727	*4*. *2*.	114	209.94	2.017		10346.	10396.	10394
1.2	401.60	2.064		788.	988.	*an .	119	206.30	2,096		10531.	10531.	10541
10	399.67	2,080		1292.	1144.	1144.	115	201.40	2.205		10811.	100'0.	19811
14	396.11	2.197		1322.	322.	1322.	110	198.40	2.140		10943.	10943.	10943
1.	393.10	2.035		1.0	1.0.	1404	110	199.15	2,083		11129	11020.	1102*
1 4	391.44	2.040		1+58.	1658.	1658.	114	1	2.077		11168.	11160.	11167.
- 21	387.48	2.096		1792.	1792.	1792.	121	1.0.93	2.099		11266.	11266.	1124.
24	343.71	2.094		1 * 32	1932.	194	144	189,-3	2.105		11476	11476.	11421
	384,74	2.063		1**1.	1941.	1991.	1	1	2.015		11466,	11466.	13464.
	141.47	2.383		71.51.	2111	2041. 2141.	1.7	144.74	2.047		11661	11461.	1166
	340,34	2.066		4410.	2750.	22 t P	121	185.41	2.104		11 '01.	11761.	11761.
2 "	176.00				312A.	2426	17	101 57	2.132		11824	11-25.	11020.
0	.,				2464		1.54	190.14	2.025		1184	11467	1104 .
N 1		2 111		2504.	250%. 2365	2564.	1.51	1. 1.	2.152		12104	12104	1210-
	172.13	2.092		2626.	2626.	2426.	134	173 10	2.074		17222	19222.	1222
	368.38	2.103		2/88.		2708.	1 1 4			2.8710	12432.	12432.	12.12
35	365,93	2.009		2765.	2763.	2965.							
\$7	361,36	1.966		5030.	3030.	3838.							
30	359.53	2.002		3234	1234	3214							
	352,00	2,836		3344.	3384.	3364							
	350.87	1.991		3649	3649.	3449.							
	376.74	1.970		5794.	1744	3794							
	344.57	2.033		3941.	1948.	3991.							
46	341.42	2.075				4093.							
• 7	335,10	2.007		* 394 .	4207. 1394.	*209.							
	334.33	2.055		**31.	**51.	** 51 .							
	189.01	2.147		4471	*329,	4339.							
31	327.24	2.849		\$791.	4791.								
	320.**	2.031				****							
	319,76	3.27		\$2.46	5146.	3146.							
	317.41	2.092		5275	*1*1.	3141.							
	315,43	1.950		\$343.	5353	5343.							
	311,70	1.990		5493.	**33.	5433.							
6 a 6 a		2.033		5600.	4606.	5680.							
		2.547		5713.	5713.	1713.							
· · ·	104.04	2.0.1		5485.	4884,	5885							
		2.020		5761.	5961.	59ei.							
**	244.24	2.010		4114.	4113.	4118.							
	296.14	8.817		6150.	4150.	4190.							
	294.94	8.014		4421	+ 764. + 371.	4321							
	240.41	2.0v9 1.999		4345.	6395.	6393.							
10		1.962		6341.	6721. 6791	#721. 4391							
	283,17	2,036		6701.	. '01,	.701.							
15	281.47	2.019		6771.	6871, 6998	6871.							
? •	280,00	2,055		7024	7024.	7824.							
7.	277.44	2.925		7082. 7198	YCe2.	1082.							
		2.0.5		7835.	,	7213.							
81	273.31	2.044		7271.	7771.	7271.							
84	212.15	2.049		7405	7347.	/347.							
**	267.40	2.049		7561.	7561.	7561.							
87	265.85	8.058		7717	7637.	7717.							
	269.25	2.058		7794	7798	7744.							
		2.945		7993.	7834, 7993,	7834. 7948.							
30	233.84	2.03/			.151	0191.							
*1	212.44	1.975		4247. 4876.	\$2%7. *178	8247.							
**	240.47	1.961			8466.	8466.							
	246.34	1.935		#235.	#535. #637.	8035.							
73	244,31 242,60	1.976		8726.	8726.	8726.							
•7		1,985		6817. 6853.	8817.	8017.							
**	240,63	1.923		8906.	1906.	8906.							
100	233.14	2.025		7826. 7133.	9026.	1026.							

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			SORIMOLE	DK **13						BOREMOLE	O# 1 - 1 3		
	PEACL	6.6	DENSITY	CURULATIVE	50L10 #885	+6#/(=++2)		PEACE	.	DEWSLTT	CUNULATIVE	50110 ##S5	1G#/[#++21
J	CLP14 DIFTI	DEN5111	OF SULIC COMPONENT	RH0=2.621	8H0#<8#0>	LINCAR SPLINC	5	D1# 11	CENSITE (64/CE)	OF BULID CONPONENT	#HU=2.821	RHDECRMOS	LINEAA SPLINE
	•		164/66)			FOR RHO				(6=/CC)			FOR RHO
											_		
	1800,86		5.8510	0	٥.	٥.	101	646,82	2,111		13552	11469.	13469.
1	409.40	2.045		•••		94 .	100	648.64	2.111		14675.	1 34 75	13671
,	908.71	2.044		192.	152.	152.	10*	6 88 , 33	2.163		13906.	13405	13906.
-		2.054		330	330.	330.	106		2.107		19117.	1-117.	14117
٠				541.	101.	541.	107	630.28	2.074		14340.	14340,	1
	896.11	2.012		873	.73.		104	624.57	2.154		1	10547.	14447
ÿ	898,20	2.096		1071.	1071.	1071.	110	421.04	2.134		14796.	1.746	14746.
14	805,40	5.875		1322	1378.	1991.	114	615.71	2.354		15113.	1-113	15113.
32	\$79.3/	2.049		1629.	1629.	1629.	113	611.50	2.724		15554.	1+134	12334
13	877.41	c.028		1744.	1744.	1841.	112	607,63	2.124		13512.	14912.	19974
12	875.66	2.061		1921	1921.	1921.	11.		2.102		13675.	1 46 7 5	12475.
1.	872.04	2.083		2000.	2048.	2000.	117	603.54	2.041		15735.	1.735.	15755.
	870.14	1.900		2224	2224.	2274.	117	599,36	2.0.7		13727	14927	12927
17		2.130		2 363	2363.	2363	150	597 97	2.073		16006.	14.704	360ru.
20	879.49	1.957		2743.	2633.	2783.	154	394.02	2.071		14401.	14201.	14201.
	805,55	100		2003.	2803.	2003.	152	591.60	2.039		16318.	1+318	16510.
23	823.18	7 127		2926.	2926.	30.4.	124	579.87	2 252		16707.	1. 797.	14767.
5	849.25	2.105		5129.	129.	3124.	13+	578.67	2.035		170.44	17094.	17544
26	447.67	2.07		3209.	3209.	3209.	151	5/6.15	2.030		17284.	17284.	17284.
21	842.75	2.19		5447	34.84	3444.	127	367.74	2.0**		17556.	17550.	17558.
29	840.78	2.10#		3991.	35*1.	3541.	130	567.67	2.028		27653.	17653.	176-3.
30	439,7y	2.111		3815	3612.	3699.	152	5.8.3.37	2.249		17849.	17810.	37849.
32	435 **	2.100		3#34,	3834.	3436 .	135	360.14	2.078		18008.	18008.	1400A.
33	\$32.31	2.053		3993.		3945.	137	339.47	2.035		18067.	1,047.	16047.
35	8.7.37	2,028		+220.	4226,	4227.	136	\$50.27	2.0**		18247.	142**	14247.
56	828.40	2.040		•277.		4277.	13/	550,/3	2.346		10474.	1.4474.	18474.
3.	419.71	2.129				4670.	139	545.61	2.044		18734	10734.	16734.
39	818,34	2,085		4702.	. 702.	4702.	140	239,70	2.044		19023.	10023.	19023.
41	814,44	2.019		5075	4075.	5074.	1.44	552.41	2.157		19311.	19333.	19397
	907.87	2.025		5204.	.206.	5204.	143	5.53.04	2.160		19483.	14483.	19463.
	804.74	2,303		1006	4363.	5506.	145	23.10	1.995		17652.	14652,	19852
4.9	798.84	2,1-1		5671.	5671.	5471.	1 **	520.01	2.056		20039.	20039.	20099.
**	796.00	2.141		5819.	5819.	5619.	147	517.62	2.107		20130.	20158.	20158.
	791.75	2.064			6042.	60.2.	39.9	513.31	2.111		20562.	20382.	203#2
	771.03	2.055			6061.	6961.	350	512.55	2.052		20472.	20422	24422.
50	783,03	2,348		****	4447.	444 .	154	500.14	2,103		20 . 37	20657	20657
54	777 96	2.041		4746.	6746.	6746.	154	505.04	2.111		20840.	20899.	20800.
	768.91	2,060		7199.	1144	7194.	150	501.14	1		20001	20991.	20991
. 55	768.12	2.899		1235.	7285.	1235.	1.76	500,42	3.002		21027.	21027.	21021
	765,76	2.111		7960	737.	7960	154		2.034		21178.	21178.	21110
48	759,85	2.165		7667.	7667.	7667.	1 3 9	493.40	2.043		21852.	21352	21352.
37	759,46	2.158		7841	7712.	7841	164		2.525		21840,	21390.	21343.
61	754,75	2.180		7	7988	7944.	168	488,90	5.005		21577	21 577	21577
64	752.37	2.100		8163	8063.	8165.	183	488,30	2.044		21345.	21191.	21545.
	748.44	2.844		#263	\$263.		165	484,56	2.033		21780.	21760	71760
\$3	746.46	2.868		6861.	8361.	8361.	144	481,41.	2.096		21937.	21937.	21777.
	739,76	2,127		#711.	8711.		1.8.0	477.94	2.043		22132.	22132	22112
64	787,14	5,103		4#14,			167	475,90	1.994		22207.	22297.	22701.
70	783.07	2,139		9967	9067.	9067	1.1	\$71.94	2.030		22395.	22314.	22319.
71	777.93	2.002		4326.	\$326.	7326.	174		2.067		22511.	22911.	22511.
	720,07	2.136		9514.	4736.	7415.	1/5	***.\$ } 464, 8 7	2.033		22:50.	22556.	22540.
,	716.11	2.149		7723.		9923	175		1.900		22168.	72768.	22 16A
15	712.**	7.154		10042.	10042.	10042.	170	457.34	2.124		- 3083.	23661.	23085.
	7 18 6 5			10314,	10174.	1031-	170	492	3 4 6		23390.	23350.	
	70			10393.	10393.	10347.	379	470.50	2.		23430.	23450.	23450.
	765,89	2.044		10490.	10451.	10441	187	445.15	2.040		3704.	23626.	23626.
	10	2.075		10609.	10604.	10609.	184		1.978		3875.	23875.	23075
82	700,98 497.44	2.327		10736.	18730.	10040.	164	436.74	2.022		28764,	23764.	23964.
	4	2.050		11006.	11006.	11004	183	4 3 5 , 5 4	2.014		24165	24165.	2-165
83	643.69	2.01*		11063.	11063.	1104".	386	432.54	2.022		24247, 34607	24297.	24247.
	6.08.16	2.044		11296.	11246.	11246.	180	449.45	1.986		24443.	24443	24
	484.20	5.014		11429.	11-24.	11424.	187	427.46	1.486		24334,	24334.	24534.
	678.74	2,079		13811.	11043.	11011.	1 7 1	474.75	2.020		29663.	24663	24663
	675.00	8.052		12077.	12677.	12077.	198	921,95 817 FT	1.966		24740.	24740.	24740.
- 93	669.66	2.030		12864	12265.	12210.	194	*15.62			23014	24079	25074
	468.09	2.002		12341.	12341.	123-1.	1 9 5	\$13.d#	1.890		25182.	25182.	27142
73 74	497.44	2.176		12072.	12672.	12 / 4 .	197		1.936		23485	23330.	29430. 29432.
	633,87	2,107		130 35.	13035.	1 5 9 5 5	, , ,		1.035		23603,	25603.	236.14
	650.86	2.075		17164. 18266.	13164.	13164.	100	397.74	2.034		25854.	25760.	25740.
100	\$49.38	2.105		13447	13347.	1 3 5 4 7 .					•		•

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			BOWEHOLE	Qu 7 - 1 3						NUMBER OF T	nu 7		
	PLACE	6 • 6	DENSITY	CUMULATIVE	SOL 10 ##55	16#/[#++21		PLACE		Dr. 413.			
	DITT	464/CC:	COVPONENT	RMD=2.871	RHC=<240,	LINEAR SP. INC	J	DEPT	CLAST	of to it	(0-0(1)14)	2.17 16 BUSE	10141
			(64/LC)			FOR RHO		D(+ +)	(6#/00)	COMPONENT COMPONENT	BM0=2.821	BHURKBHC /	51 14. Fta A-
294	394,30	2.050		25750.	25950	25940	ť	1800.00		2.+210			
200	343.20	2.717		26083	76083	26043.		1942.41	1 1 0 0		¢.	р.	
105	190.03	2.916		20137.	26139.	26199.		1041.86	7.113		165.	145.	14*
206	388, <i>81</i> 186, 94	2.024		26268	24248	26288.	;	1040,00	2,083		246.	246.	2.4
20+	346.11	2.044		26384	26384	26344.		1887.30	2.090		303.	385.	385.
207	382.76	2.064		24580.	26580	26560.		1034,77	2.991		526.	526.	526.
211	542.17	2.05+		26600.	26600	26600.	¥	1035.20	2.030		604.	60.	404.
515	380.20	2.050		26716.	74716.	26716.	11	1020.01	2.000		/55.	7-5.	795.
21.	374,67	2.052		26786.	26794	26744	14	1024.90	2,000		903.	403.	903.
215	373.90	2.052		27024.	21024.	27024.	13	1924,54	1.442		1418.	1010.	1010.
117	367.76	2 893		27239.	27139.	27144.	12	1017.45	2.094		1454.	1344	1999.
210 219	366,01	2.000		27367	,,,,,	2736-	1+	1015,40	2.040		1953.	1453.	1452.
110	362.55	2.044		27566.	27*61. 	27566.	1.	1004.30	2.060		1744.	1428.	1.020
111	360.31	2.049				27683.	17	1005.84	2.6 4		19-1.	••1.	1941.
225	129,37	1.936		27831.	2*8°1. 27917	27845	21	1000.13	2.07		2421.	2741.	2221.
27	3*3,C*	1.99*		28024	20 24	2007-	24	998.30	2,040		2300.	, 100.	23-0.
	447,51	1.95		28134 2827#	20134.	28144.		97. 48	2.0		2393.		
221	540,42	2.069		28613.	28613.	28613.	23	***	1.985		2.41	2541.	23+1.
229	333,44	1.944		20726.	28726.	28726.	27		2.050		2123.	2650.	2121
230	334,54	1.786		28886	28886.	20546.	20	986.74 885 VL	2.035		28.94	2854.	28%4
232	333,73	2,030		28922.	28922.	28922	30	982.81	2.104		2893. 304c.	2893. 1050	2893.
235	529,00	2.031		291+8.	24144.	29348	31	981.23	2.086		31 *1.	\$1.51.	51.51
133	328.44	2.020		29185.	24185.	29185.	33	976,11	2.002		3272.	3772.	3272.
236	322.70	2.061		29450.	2443D	29090.	34		2.124				3***
238	317,19	2.078		29588.	29580.	29388.	36	970.21	2.017		5816.	3616.	3616.
237	315,82	2.867		27823.	24423.	27423		967.04	2.035		5816.		3016.
2+1	309,31	2.004		30018.	30010.	3001A.	37	** 3 . **	2.074		3734	3855.	3854.
242	306.95	2.020		50225.	30225.	3022*.		962.53	2.110				
244	303.41	2.030		30344.	30 300.	90996.		979,82	2.063		**85.	4761. 4483.	4261.
243 246	302.62	2.002		30432.	10= 12.	30417.	• • •	953,24	2.030		. 360.		4360.
241	249.41	1.939		30360,	30560,	30360		920.13	2.047		112	\$712.	4712
248 /47	296.32	1.936		30716.	30716.	10734		948.14 947.97	2.013		4807.		
250	291.59	1.937		38727	30927.	30096.		945.40	1. ***		**36.		4136
124	584 40	1,955		10744	32446.		5.	9-1.00	1.947		3046.	5048.	5048.
(53	241.74	2,036		31 46 5	1 365	31365.	21	416.11			52.2	3252.	2252.
157	275.04	1,042		31486.	31486.	31486.		936.30	2.090		5350.	5350.	535
25.B	275.01	2.017			31646.	114=4.		3	5.085		5491.	SH21.	5451
3.	270.34	2.020		51778.	31778.	31774.		946.04			7710. 5048.	9510. 5688	551C.
237	267.17	1. *25		31**5	31995.	31795.	21	336,44 9,8 4/			3468.	****	5668.
264	258.91	2.053		32246	32236.	\$2236.	5*	9.6.08	2.044		5'84.	4766.	5764
che cha	244,72	2.111		1	12999	325.4.	FU 51	142.33			1923.	\$923.	2923
***				32940.	32694.	32644		10.0	7.110		135	4105.	6101.
263	243.44	1.497		\$3170	35170.	33170.	6.5	919.03	2.440		6160.	6168.	6168.
	238.44	2.047		34419.	33226.	83226.	63	417.06	2.305		6355.	6232.	4352.
164 164	234,88	2.053		33593.	33593.	33993.	67	915,04	2.079		4456.	6456.	6456.
79	888.50	2.000		38789.	33710.	33710.	68	910.7b	2,105		467A	6678.	6536. 6678.
74	225.44	2.034		3****	3.026.	34826.	70	909,30 906,00	2.125		6740.		
73	221,00	2.038		15214.	34234.	34836.	71	90	8.1*5		7009.	7601	7004
78	220.51	1.473		3*312.	34317.	34317.	7.5	404.00	8.140		7030.	7030.	7030.
17.	\$24.14	1.976		31541	34349.	34344.	7.	499,75	8.097		7460.	7260	7260
78	572.01	1.8476		34620.	34670,	34820.		676.17	2.091		7.01.	1401	1001.
7	207,11	1.918		54877	34877.	34877.		845.41	2.127		7505.	2583	7503
	103.3	3.426		33870.	33070.	850T6,	77	888,32	2.050		7684.	7684	7684.
95	102.11	2.136		34114	36139.	33746,	E U	885,96	2.151		7962	7962	7962
	189.28	2,190		16157.	36157.	46197		882.44	2.145				
83	,	•••	8.8810		36143.	86145,	63		2.100			8255.	
							85	817.44	2.113		8337	8337.	
								876.51	2.179		6-5A.	A438.	8×38, 8×98.
								#75,34 871,78	2.040		8558.		8544
								873.51		2.6210	0699. P712.	8649. 8712.	8699.
							40	871.30 870,44	2 840	\$, \$255	*13.	8713.	• 7 1 2
							92	479,21	2.050		8738. 8775	8737. 8775.	8774
							4.5 74	867,46 866,84	2.290			6 9 2 3	
								865.98	2.010		9912.	8935.	8735.
							96	###.71 ##1.33	2.000		90-6	9069.	9049.
							21	835.94	2.110		9966.	**52	9442
							100	828.48	2.09%		9327.	9512.	9512
												1601.	

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			310H3#C8	0+2-18						804LHOLE	0#2-18		
	PLALL	6 • 6	0245111	CUMULATIVE		(6#/[*++2]		PLACE	6 - 6	DENSITY	CUMULATIVE		
J	0181) 0181)	DENSITE SENJECT	0" 50610 CONFONENT (84/CC)	BH085-851	Rupecandy	LINCAR BPLINC For AND		DILI	DENSIT 194/201	CDAPONENT (SA/CC)	#H0=2.821	RN0=(#H0)	LINEAP SPLINE FDP RHE
	852 ZU		2 9130	7690.	9623.	1623.	201	627.97	2.129		19465.	19405	19905.
104		8.047		7801.			202	650.40	2.141		195.4		17467.
10.0	846,87	2.866		9917.	4893.	10044	10-	691.70	2.110	2.7303	19998	19823.	19823.
10.		2.099		1841.	102 .	1.0244	286	487,47	8.124		28124	28075.	20075.
100	437,14	3.074		10175.	10343.	10363.	200	698.23	2.110		20207.	20139.	20154.
100	633,37	1. 486		10360.	[n333.	10333.	230	\$*0.27	2.144		20502.	204 99.	20-54
109	652,90		2.9292	10603,	10564.	10564	214	637,78	2.204		20702.	20195.	20345.
		3.905		1971.	10676.	19676.	23.4	433,99	2.704		20859	20424	20824.
112	P24,00	2.000		10825.	10783.	10783.	212	632.00	2.129	2.1243	21012	20499.	20959
13.	923.15	2.035		11027.	10903.	10983.	21.0	630.05	2.126		210-3.	21043.	2104
115	223.20		5.9500	1105%.	11010.	11010.	219	628,87	2.044		21135.	21106.	21106.
117	819.01	2.108		11223.	11176.	11174.	811	\$22.57	2.110		21457.	2 . 4 3 4 .	21434
110	N10.44	5.089		11304.	11255.	\$1255.	514	621,39	2.151		21519.	21498.	2144P.
120	813.6V		1.9383	11341.	11593.	11+87	420	636.00	2.192		21777	2174	
121	813,12	2.002		11565.	11511.	11511.	221	616,40		2,7201	21792.	21 / 77.	21777.
124	811,15 809 uu	2.125	2.9043	11667.	11611.	11771.	223	631.9*	2.1**		22038	21875.	22929.
124	AR8.74	2.046	••••	11789.	11751.	11731.	224	610,36	2.11#		22123.	22116.	22116.
123	BU7,41	2.097		11067.	11810.	11810.	225	606.03	2.045		22286.	22282.	22282.
127	804.07	2.07		2020.	11744.	22944	227	603.00	2.027		22477.	22477.	22477.
120	802.49	2.07		12107.	12044.	12044.	228	685,30		2.7140	22514.	22515.	22515.
130	774.19	2,126		12434	17369.	12365	234	599.35	2.101		22668	22672.	22672
131	794.00		2.91#3	12347.	12476.	12476.	231	575.80	1.808		22832.	22840.	22840.
134	791.06	2.06		12032.	12502.	12458	233	943,50	1.647	2.7613	22928	22922.	22936.
134	787.74	2.035		2846.	12770.	12770.	234	549.07	2.140		23101.	29111.	23111.
1.50	787,14	2.047		12085.	12007.	12407.	239	503,39	2.144		23223.	25234.	23234.
1.97	785,58	2.014		13417.	12937.	12937.	231	582.91	2.212		23443.	23502.	23592.
158	182.41	1,999		13109.	1 5028.	13020	234	580,05	2.097		23623.	23628.	23628.
100	778.87	1,907		1 3276.	15173.	13191.	240	578.90		2.9136	23680.	23686.	23686
348	775.78	2.036		13425.	13337.	13357.	547	576.90	2.047		23179.	23784.	23784.
144	773.00	2.104		13343.	1 1 4 5 4 .	11454	244	573.30	2.189		23969.	21970.	23970.
1.44	115.10	2.000		13624.	1 1532.	13532	244	370.99	2.454		24104.	24103.	24104.
144	778.60	2.06*		13423.	14591.	13776	244	343.07	2.133		24326.	20322.	24322.
147	767.45	2.082		13011.	13745.	11745.	241	563.91	2.075		24986	24480	24480.
1.44	766,90	2 033	2.9706	13868.	11772.	13772	248	579.30	2.003		24645.	2+638.	24638.
190	763.04	2 044		3950.	13861.	13861	234	354.90		2.4723	24/39	2. 730.	24730.
171	762.33	8.064		14096.	13995.	13995.	231	558,79 556 us	2.060		24744,	26736.	24736.
1.94	738.79	2.114		14278	14175	14174	254	553.0/	2.1.30		25410	24996,	24946.
194	737.61	2.140		1.3.0.	14236.	14236		535.10	2.105		25043.	24080.	ZSORC.
150	754,98	2.140		14526.	14317.	14420	215	5+8,35	2.030		25271	25121.	20171
191	192.39	2.079		14627.	1+519.	14519.	23/	546.27	2.038		25366	25351.	** 3*1
134	747 11	2.105		1-051.	14739.	1		548.04	2.035		23542.	24426.	27426
1.0	146.37	2.183		14913.	1.799.	14744	260	241.47	2.090		25621	24682	23602.
101	741.47	8.191		19181.	14064.	19044	164	523.10	1.124	2.8882	29927	2000	23704.
18.0	1-1.01	2.190		13402.	15085.	12085	263	532.00	2.151	•••••	26116.	26041.	24091.
142	739,69	2.15/	2 8837	15443.	19148.	15243	26.7	528.87	2.132		26286.	26257.	26247
1	786.14	8.097		15482.	15312.	13312.	74.6	575.15	2.134		26539	26509.	26509.
167	738,99	2.097		15574.	19452.	13452.	267	521./*	2.099	2 86.43	26664	26633.	24633.
167	732.04	2.127		15675.	15555.	19593	267	516.66	2.140		e6952.	26897	24899
170	728.4/	2.058		15857.	14755.	1914	270	515.30	2.110		27056.	27022.	27022.
174	722.96	2.151		14135.	16015	16015		510,74	2.145		27243	27200.	27208
173	720.99	2.145		16841.	16125.	14173,	273	408.79	2.168		27450.	27314.	27.51.4
175	719.00	2.137	<i></i>	16.54 7	16211.	14431		505.07	2,006		27547	27560,	27560.
174	715.64	4.114		164 .	16349.	16359.	276	501.51	2.000		27751.	27694.	27494.
178	712,33	2.197		16843.	16737.	16737	270	478.55	2.065		27964,	27827.	27827.
114	109.20		2,7303	16963.	16758.	16748.	277	476.17	2.085		27983.	27946.	27946.
1.00	707.41	2.149		17056.	16843.	16754	284	493.40	2.044		28922.	27985.	27985.
184	748.87	2.1.0		17162.	17063.	17063.	284	\$92.65	2.124		20160.	20123.	24123.
184	678,93	8,160		17417.	17325.	17323.	28.3	491.90		2,8079	28199.	28163.	20163.
185	675.80	2. 223		17591.	17500.	17500	285	490.07	2.118		28285.	28248.	282.8
184	675.40	2.209		17613.	17523.	17525.	286	489.10	5.140		28348.	28312.	20312.
180	689,50	8.126		17940	17854.	17656	280	183.76	2.114		28516.	28479.	20119
189	686.74	2.145		18087.	18004.	18006	287	485.48	2.113		28598.	24561.	28561.
194	681.84	2.053		18564.	18068.	18584	241	479.26	2.205		20747. 20480.	28709. 28850.	20709. 2084C
192	680.04	2.099		18344.	18510.	18310.	294	478.87	2.71*		28903.	24862	22862.
1.44	679.147	2.093		18305.	18453.	10433	293	476.51	2.157	2	29036.	28992.	28992,
199	672.50		2.7221	18822.	18756.	18746	295	474.54	2.116		29139	29095.	29095
176	669.81	2.088		18961.	1.498.	18898.	296	472,96	2.050		59574	29173.	87173.
	662.47	2.082	E. * 1 * 8	19201.	14137.	19139	270	***.21	2.047		29449	7923U. 29401.	29401.
197	\$43,91	2,0		19262.	14499.	17149	274	*****	2.074		27347	29497.	29497
100	901.00	5.105		17383,	19322.	14.25	300	483,48	2.050			29536.	29536.

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			BUREMOLI	0#2-18						804(#OL(KA#-01		
	PLALL	6 . 6	DENSITY	CUMULATIVE	40:10 HASS	164/284421		FLACU	5 - F	PENSITY	CUMULATING		
J	DIFT:	0645117 (68722)	OF SULID (DAPONENT (GR/CC)	#H0#2+#21	840#K849>	SPLINE FOR RHC		DIFI	CEN5233 16#/CC3	OF SULID COMPONENT IGN/LC)	840z2.821	8-0 2(8 43)	<u>, INE</u> 64 50, INE 6,0 040
01	-62.73	2.850		29722.	74670.	29670.	U	1800.00		2,8110			
104	461.94	2,020		29760.	29707.	25707.		1185.85	2.085		n.	0 .	¢.
	***	2.147		38977.	30082.	10022.	5	1182.48	2.057		146.	1.40	145.
185 185	454,87 858,88	\$130		30163. 10296	30107.	\$0107. 30170.	:	1101.47	2		297	258.	210.
101	*52,84	8.197	•••••	30227.	50171.	39171.	•	1111.41			375.	\$75.	375.
104	438,77 448,16	2.115		50 832.	30775.	30414.		1175.44	2.044		\$34.	534.	334.
		2.124		18732.	30671.	30671.	•	1178.19	2.044		632.	632.	632.
611 614		2.157		50836	30774.	50774,	11	1366.17	2.042		918.	918.	
	457.14	2.193		31053.	394#7.	30987. 31177	14	1144.14	2.025		101*.	101	101
11.2	438.92	2.957		31 390.	31231.	31231.		1159.34	2.045		1249.	12**	12.4
816 817	440,44	2.027		51396.	31327.	31375. 31301.	17	1158.13	2.017		1917.	1911.	1452.
1.	-281	2.279		51*91.	31919.	31417.	11	11 42 . 93	1.996		176*.	547	1547.
22V	425.74	2.090		31432.	31559.	21759.	1	1120 40	2.090		1744.	1744	
121	419.81	2.871		\$1 9 31.	31856.	11856. 11882.	24	1100.04	2,070		1004.	1804.	1804
121	418.65	2.002		31990	31916.	51916.	24	11-1.4	2.974		2009.	2009.	10.05
325	*13,31	2,012		32246.	37171.	32171.		1142.46	2.042		2040.	2040.	2045
\$2.	101.30	3.984		32331.	57355.	32344.		11+0.44			2149.	2149.	/149. 2282
321 370	*93,45	1.946		\$2624	32548.	12548.	2	1156.4-	1.978		2351.	2357	23.1
327		1.915		32/26.	32649.	32649.	2	1134.73	2.041		2468.	2468.	2468.
131	377.34	1.939		32901.	32802.	92002	3 0	1130.01	2.034		2641.	2627.	2
334	5*7.3/ .**	1.011		32968.	3288A. 19974	32888. 32974.		1123.90	1,975		2908.	2808.	2941
	\$ 7 5. 90		2,4045	35124.	19011.	530+1.		1141.97	2.055		50a1.	4041.	3041.
335	349,49	1.444		33308.	\$\$222.	33227.	35	1110.76	2.015		8195.	3195	3145.
33/	376.40		2.8478				30	1116.75	1.494		5289.	3289.	3289. 3423.
337	147.40		2.0768					1115.33	1.980		3497.	4477.	3+49.
540	531.1V 513.9V		2.4745				37	1110,52	1,980		3391.	1591.	3591.
344	244,90		2.8892				•1	1107.91	2.053		5701.	.701.	3701.
343	217.90		2.08793					1106.71	2.047		3061.	3861.	5861.
345	249.90		2.0014					1101.07	1.967		3996.	3996.	3996
	232.99		2.6210				**	1048.41	1.966		.15.	.154.	*1 **
3.8			8.4510				4.7 5 0	1097.47	1.978		4251.	6196. 8751.	4146.
								1893.00	1.974				4306.
							54	1098.47	2.043				
							34	1090.44	2.049		*525.	4525. .708	
							59	1003.04	2.070		. 165.	.765.	. 76 .
							55	1083.81	2.081		4968.	**?*	4868.
							21	1001.40	2.120		4441.		• • • • • •
							57	10/7.70	2.202		5168.	.15.	5166.
							61	1076.30	2.224		5257.	5257.	1271
							62	1075.75	2.106		5*1*.		5-1
								1048.7	2.010		5650.	5650	54 SC .
							45	1066.55	1.994		5763.	5763.	1743.
								1060,10	2.010		6073.	6073.	6071.
							.,	1053.00	1.970		4173.	6373.	6375.
							70	1052.47	2.010		6507.	6429. 6907	6429. 65n7.
							74	1040.05	2.025		6643.	6442.	
							15	1045.64	2.034		6759.	6720.	6790.
							??	1045.60	2.07*		6849.	6459.	6859.
							1	1047.61	1,984		7142.	71.2	7147
							7.	1057,21	1,978		7160. 7215	7160. 7285	7160.
							Ru	1033.54	2.018		7351.	7331.	7331.
							81	1030,50 1028,37	2.143		7598	7993. 7998.	7545
								1025.75	2.0**		7 7 7 8 .	7730.	7716
							85	10-0.95	2.0.4			7920. NADL.	8001.
								1017,52	1.990		8133.	M1 13.	8133. #242
								1013,30	2.1*8				8337.
							87	1011.49	2,138		8466.	P.44.	2346.
							91	1000,20	2.026		6603.	8603.	8403,
							93	1003.00	1.947		8875.	8875.	8875.
							94 44	995,64 990.60	2,106		9241. 9873.	4241.	9241. 9475.
									2.1.17		9557.	9557.	9537.
								979,85	2.230		10094	4716.	10046.
								976,74 975 AF	2.130		10223.	10273.	10223.
							,					1030.01	

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			BUREMOLE	# AH - 01						BOREHOLE	###+01		
	PLACE	6 - 6 06 #5177	0545171	CUMULATIVE	50_30 #455	16#/[#++2]		PEALL	s - s	3645271	CUMULATIVE		(6#/(Hee2
	D1F1)	16m-221	COUPONENT IGN/LCI	AH0=2-821	RHU& (8HD)	PLINE FOR RHD	5	DIFTI	(64/CC)	DE SULID SDAPONENT (S4/SC)	NHU22.821	RH08(840)	LIVEAR SPLINE
													· UK KHU
105	969,31	2.045		10466.	10466.	10606.	202 202	749.36	2.071		21756.	21756.	23746.
10.	967,3V	2.041		10709.	18704.	18709.	283	795, 29	2.090		21962.	21879.	21879, 219,2.
142	938,86	1,761		11128.	10474.	11120.	205	743,93	2.045		22043.	22043.	22841.
100	955,65	2.191		11286.	11206.	11204.	500	791.14	2.0%1		22184	22104	221C . 4218*
108	999.44	2.042		11453.	11513.	11013.	290	789.11	2.0/4		22/85.	22285.	22281.
109	948,44	2.146		11655.	11655.	\$1655.	297	732.60	2.101		22619	224 2.	224 17.
- 111	912,00	2.157		11966.	11944.	11966.	214	727,47	2,130		22787.	22787	22767
112	240'90 247'22	2.132		12010.	12010.	12010.	234	725,85	2.127		22978	22974	22978.
114	937,57	2.1*6		1.245	12245.	12245.	213	723,83	2.122		25020	23020	23020.
115	933,36	2,101		12830.	12338.	12336.	\$13	728.65	2.098		23246	23244	29246.
117	981,22	2.396		12563.	12563.	17363.	21/	717,42	2.090		23348	23348.	2334ê. 2380ê
117	748.15	2.107		12712.	2269D, 13712,	122490	210	715,81	2.133		25495	21493	25441
750	969.76	2.124		12929.	12979	12929.	220	712.60	2.12		23562	23578.	23178,
124	941.14			13118.	13035.	13833.	221	711,59	2.130		23727	25727.	25171.
123	918,47 917 00	2.154		13265.	1 1265.	13265.	\$53	709.50	2.114		23833	23791.	22793.
143	915,40	2.1 %		14414.	13419.	13919.	225	705.36	2.05		23937.	79937	23917
124	414.67 912 bb	2.106		13462.	13462.	13462.	226	704,30	2.040		24679	24079	24074
124	911,06	2.0.6		13644	13644.	13644.	228	708.36	2.100		24140.	24140	2+1+0.
1.50	999,85	2.104		13706.	13796.	13706.	554	6 78 . 74	2.100		24368	24263,	24364.
131	985,83	2.127		13759.	13959.	13959.	527	698,13	2.074		24404.	24404	2.404
130	904,23	2.151		14067.	1-002.	14802.	254	692.70	2 015		24725.	24725.	24774
1.34	901.01	2.202		19178.	1+178.	19178.	234	698.94	2.018		24763.		24763.
135	877.40	5,100		14556	18373,	14373.	235	687,64	2.033		24919.	24719.	24939.
131	873,30	2.042		19578.	14578.	14578.	237	601.60	2.078		23074.	24074	25074
134	809,76	2.796		19741.	3+698.	14648.	230	680.86	2-1-1		25495.	25275.	25245
1.44	845.75	2.037		14963	18963.	1	244	676.44	2.050		25337	24357.	25357.
142	880,93	2.106		15203.	15001.	19001.	241	673.eż	2.049		23439	24639	23639
143	#77.J1	1.993		15461.	15361.	15361.	2=3	669.61	2.041		25/21.	25771.	25721.
145	874,98	2.015		15092.	19992.	13492.	244	647,8U 665.37	2.008		25971.	25971	2 2 9 7 1 .
146	872,49 870.88	2.015		15367.	15587.	13587.	246	663.70	2.047		26046	26019.	26019. 26098.
1		2.045		15004.	15804.	13404.	244	639,37	2.025		26215.	26215.	26715.
124	887,85	2.100		15685.	15485.	15885.	247	658,76	2.025		26349.	24349	26311.
191		2.101		16030.	16030	16030.	231	638,35	2.047		26466.	26466	26466
199	850.44	2.625		16135.	16135.	16135.	232	692.74	2.047		26643.	26643.	26603.
15*	856,92	2.045		16396.	16396.	14896.	234	649.70	1,985		24720. 36796	24720.	26720.
130	874,84	2.095		16537.	14937	14455.	235	647.31	2.025		26888	26888	26480
151	851.60	2.058		16658.	14458.	14458.	257	642.69	1.994		27061.	27061.	27061.
139	844.47	2.117		17018.	16774.	16774.	254	638,67 683.86	2.033		27308.	27508	27308
363	871,96 839,98	8,802		17137.	17137.	17197.	360	632.65	2.025		27595.	27543.	27541
164	837,94	2.274		17355.	17333.	17515.	464	640.44	2.010		27670.	27670.	276.70.
164	835,93	2.095		17394.	17344.	17344.	26.5	628.65	2.241		27784.	27784	27784
165	\$55.20	1.999		17572.	17572.	17572.	269	624.21	2.019		27881.	27061.	27881
1.57	827.49	2.042		1785.	17753.	17753.	26.0	622.60	2.07.			28080.	20080
164	874.60	1.746		17967.	17987.	17947.	260	418.99	2.090		20263.	26101.	28181.
174	\$20.25	1.903		18180.	10077.	18477.	267	617.70	2.095		\$4425.	20323.	28175
171	817,45	2.010		18862.	1.242.	18242.	271	613.06	2.098		24351.	29427.	28421.
173	414.43	2.0%5		18358.	18858.	10390.	27.5	689,43	2.074		28673.	24673.	20673.
175	811.74	2.007		18477.	18477.	18477.	274	607.76	2.042		28833.	20033,	20033
176	809,01 607 81	2. D		18728.	18724.	14724	274	602,74	2.170		29010. 29073	29010.	29010.
174	805.80	2.124		14844	18789.	10707	271	£00,Y1	2.227		27185.	29185.	29185.
179	802.37	2.100		19061.	1 .061.	19061.	279	576.87	2.140		27819.	24319.	29319.
141	797.21	2.108		17211.	19206.	17204.	590	593.20	2.153		79598.	24598	27596.
183	795.85	2.111		19436.	19436.	19936.	284	584,44	1. ***		27948.	24848. 24989	29848.
194	791.74	2.098		19621.	19517.	19519.	283	582,93	1.978		30116.	30116.	30116.
182	786.92	2.00/		19745.	19765.	19745.	280	5	2.100		30466.	30365.	30464
187	105.31	2,977		19945.	14945.	19905.	284	573,19	2.071		30544	30.548	30.544
199 199	775.21	2,030		20233. 20646	20233.	20213.	284	569.17	3.959		30745.	30629.	30629 . 307es.
199	775.26	2.023		20553.	20353.	20555.	540	566,76	2.025		30896	30856.	30836
745	764.64	2.031		20611.	2:	20611.	271	560./*	1.878		\$1125.	31125.	31124
143	767.60	2.325		20 . 0	20430	20 . 0	295	595.20	1,967		31286	31246.	31246.
195	761.61	2.101		21054.	21034.	21046,	794	993.94	2.095		31732.	51532.	51532.
1.46	759.60 758.44	2.098		21240.	21240.	\$1540.	276	547.00	2.149		31680.	31680.	31600.
190	156.19	2.114		21902.	21302.	21302.	297 234	546.67	2.116		33 /06	31784.	\$1786
197	754.88	2.093		21710.	21.10	41910.	299	545.00	2.196		3)892. 31776	32892.	31842
				44 0 75.	21613.	21413.	56 U	540.67	2.100		32101.	32101	32101

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			8 38 E HOL E	#48-01						904L 40L T	***-63		
	-		DENSITY	CUMULATIVE	\$0111 -453	15# 1240021		Plat	3 - 3	TENSITY	COMULATIVE	ROLIE NASS	
5	01.41	184/111	154/LL)	#HU#2+823	847.8CRHU4	LINEAN SFLINE Fjar¢Htj		2001	1	16 80 10 16 87 01	LH082-823	₽₩₽₽₹₽₩₽₽	SF, INF FUR RHY
301	347.43	2.058		32264.	12264.	32264.		314.98	1.844		-2307.	*****	42501.
302	535 63	2.014		32342.	32	32342.	- 1	316.87	1.454		• - 71	+ 2 5 7 1	
303	534.44	2.010		32999.	32349.	32495.	40.4	309,82	1,8*2		42420,	42420.	42420.
305	531.41	1.999		32389.	32949	32549,	4.75	305.43	3.01*		42.84	\$2784	
304	528.17	1.996		32642.	32642.	32642,	406	304.02	1.070		42651.	428%1. 44870	428411
300	527.59	2,010		32 1 14 .	39736	52136.		303.00	1,991		.2088.		
314	526.17	3,978		32743.	32793.	37743.		301,21 300.81	1,855		62973. 62989	42973.	42975.
311	525, 31	1.951		32916.	32416.	32916.	•11	300.41	1,867				
314	348.17	2.050		33445.	32970	32990.	414	249,60	1.634		45034	41059	*3039.
314	518,92	7.001		33123.	2 3 1 3 .	\$1173,	•1•	293.70	1.80%		+ 3256 .	* 3256	+ 3256 .
319	516,97	2.134		33872.	33222.	33292,	*17	241.1/	1.070		43844.	*****	43344,
311	913,13	2.058		13587.	33387.	\$5887.	•17	287.15	1.8**		43429.	* 3 * 2 * .	
214	505./0	1.871		33384.	31784.	33344.	* 1 *	201.33			41445.	41495. 41578	43445.
5.0	504.49	1. 150		33793.	31793.		420	284.34	1.07-		. 3620.	43678	43628.
324	\$99.61	1.471		33843.	33883.	33968.	421	281.14	3.010		43761	* 1761.	
\$23		1.648		34824.	3.024.	34024.	*23	278.12	1.870		+ 3867	+ 3862	1386.
325	495,82	1 9 3 2		34306.	1410 4 .	34306,	• ? •	276,74	3,970		*37*3.	624¥3, 6.007	43963. 46071
326	4 9C . U.S	1.870		S#280.	3.280.	34280.		214.64	1.554				
340	488,92	1.824		34397.	34327.	44327. 34436	*2/	272.48	1.860		44103 .	N#193.	
124	483,60	2.234		3*363.	34963.	3=563.	127	2.9.47	1.739		44247.	** 2 * 7	
538 533	481.99	2.079		39649.	34649.	34669,	4 5 0	267,86	1.711		44287.		142F'.
334	477.YB	2.390		34861.	34861.	34861.	• 3 4	265.44					
335	473,9/ 473,96	2.010		34959.	34959.	34959.	• 5 3	263.04	1.803		***58.	***58.	***58.
335	\$72.35	2.242		35151.	35151.	35151.	• 3 5	241.44	1.840				
330	471,37	2.154		35218.	35218.	33210.	- 34	259,43	1,827		64597	** 597.	44547.
3.58	467.13	1.899		\$5390.	13390.	35398.		274.44	1.766				*****
337	*66.33	1.855		\$3922.	39926.	33992.	* 3 7	234.20	1.759		**7#3.	** 783.	
143	462.75	2.0**		35564.	39* 88 .	35568.	***	249.19	1.629				
313	438.47	2,007		35405.	34707.	35405,	***	248.58	1.675		44964.	44964.	44964. 49015.
***	457.09	2.0.**		35863.	39863.	35863.		244.96	1.00				
345	\$32.61	2.034		35760.	34460. 86379.	11 76 0, 36079,	**5	248,76	1.711		45126.	4-126.	45176.
347	451,46	2.100		36167.	36162.	36162.	447	295.19	1.511		45231.	45251	.5231.
344	847.04	2.076		36364.	36764	36246,		239.14	1.548			**2*1.	45241.
350		2.154		36-66,	36.466.	36966.	150	236.33	1.961		45.524	****	45324
554	443,82	2,090		36243,	36747.	36343.	424	254.13	1.564		4533".	45335.	41114
\$7.5	439,43	1. ***		36763.	34763.	36763.		2 30, 30	1.742			45493	45-43
374	436.49	1.673		36873.	36073.	36973.	* * *	229.34	1.669		\$3533.	44533,	43533.
554	\$32.18	1.8*6		37037	\$7:57	37037.	456	203.00	1.569				44492
357	426.17	2.202		37191.	37991.	37338.	130	220,06	1,908		45F31.	45851.	41431.
359	423,74	2,982		\$7457.	57=57.	37957.	+37	217.65	1.5+8			45*11.	45911.
361	418.74	2.196		37651.	37966. 37681.	37681.	460	215.64	1.552		45960. 5600b	45960	43460.
34 4	\$15,74	2.030			37448.	37848.	46.0	211.04	1.708				
363	414,90	2.138		30076.	34887.	30076.	463	210.41	1.783		46111.	44111. 84125	46111.
365		2.042		38238.	38730.	10236.	463	709.61	1.742		¥4155,	46153	+6153.
364	404,00	2.042		38295.	38276.	34349.	766	209.21 208.0v	1.755		N6167.	46167.	46167.
364	483,65	2.122		38963,	38463.	38463.		206.40	1.659		++263.	46263	46263.
578	376,85	2.138		38492.	38678.	38492.	478	204.14	1,625		44311.	46311.	46311. 46358.
372	\$97,25	2.339		38786.	38786.	38786.	474	201.90	1.612		46392	16392.	46392.
57.	394,00	2.334		37985	34043.	37065.	173	198.76	1.699		46435.	4643".	46476.
\$74	344,54	2.138		37825.	39725.	39225.	. 7.	197.15	1.931		46536.	41.536.	46516.
	303,97	2.122		37304	39506.	39506.	476	199.75	1.915			46407.	44607.
377	382,76	1.974		39865.	39565.	37565.	477	192.19	1.760			*6721.	46721.
979	337.43	1.924		.0676	40676.	49676.		109.73	1,981	7.8710	46731.	46831.	*6031.
184	375.44	3.899				40744.							
384	372.00	1,985		10767	48767.								
3=3 586	348,21	1.876		*1936.	41436.	41836.							
387	3+5,80	1.827		41138.	41133.	41133.							
384	341.79	1.817		41246	+1286	*1286.							
387	337,31	1.711		41386.	41386. 41443.	*1306.							
387	356,18	1.787		41487.	41487.	*1*#*,							
391	332.24	1.739		41971.	41571. 41411.	41371. 41611.							
194	330.13	1. 977		41707	. 707	.1707.							
394	326,75	1.979		41778. 41849.		41778. 41844.							
395	324,91	2.131		41749.	.,								
347	319.49	1.849		\$2899. \$2245.	42049. 64283	420 4 9.							
398	\$10.90	2.010		+22+1.	*27*1.	+22+1.							
40 H	514,0/	1.972		42334.	\$2335. \$3887	42334.							
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FIGURE 6-29. -- Change in rock thickness from Y-Y densities, assuming simple Borehole ORT-20 vs. OAR-2A. subsidence.

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FUGURE 6-40. -- Change in rock thickness from Y-Y densities, assuming simple subsidence. Borehole OIT-11 vs. OSR+21.



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CHAPTER 7: <u>INTEGRATION OF MATERIAL-PROPERTY UNITS, GRAVIMETRY,</u> <u>AND ADDITIONAL STUDIES OF OAK AND KOA CRATERS</u>

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INTRODUCTION

Preliminary interpretations of the geology of the OAK and KOA crater areas and of the craters themselves are presented in Wardlaw and Henry (1986a, 1986b). Since those reports, additional information was developed from analyses of borehole gravimetry, paleontologic mixing, thinning, and distribution of shocked calcite, most of which are presented in previous Chapters of the current Open-File Report. These new data require modification of the geologic interpretation of OAK and KOA craters. This Chapter incorporates these new salient data and presents a more comprehensive interpretation than that given by Wardlaw and Henry (1986b). Depths to a few horizons or zones have been reinterpreted, and all pertinent data are presented herein in corrected form as tables. These data supercede all previous information.

The most convenient way to relate the geology to crater phenomenology is to develop geologic material-property units that match the general materialproperty models for OAK and KOA craters. The geologic framework is reviewed briefly before presentation of the new geologic material-property units (MPs). These units will be used throughout this text in deference to previously used geologic schemes such as the sedimentary packages (SPs) of Wardlaw and Henry (1986a).

PRE-EVENT GEOLOGY OF OAK AND KOA CRATERS

The general stratigraphic sequence of Enewetak Atoll is punctuated by a series of discontinuities within the carbonate sedimentary rock column, of which uine are identified as major disconformities in the upper 1,200 ft (Wardlaw and Henry, 1986a). These major disconformities represent significant exposure and cementation surfaces over most of the atoll. Generally, pervasive cementation is confined to the reef margin (fig. 7-la), but extends for a considerable distance beneath the lagoon beneath disconformities 5, 8, and 9 (fig. 7-2). Data from the EXPOE Project (Couch and others, 1975), which presents data from shallow boreholes drilled on islands on the reef tract, indicate that the geology is generally similar throughout the reef tract (fig. 7-lc), although the width of the cemented reef margin narrows on the leeward side of the atoll. Cementation also appears to generally decrease in areal distribution in the sequence from disconformity 5 (Pliocene) to disconformity 1

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disconformities (1-9), biostratigraphic zones (AA-MM), material property units (MP-1 to MP-5), and sedimentary packages (S^p1 to SP8) in reference boreholes. Highly organic zone (SP5) is stippled in columns. FIGURE 7-2. -- Relationship of discontinuities (shown in columns), major

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(Pleistocene) as represented in the cross section of Figure /-.d. From disconformity 1 to the present surface, the area of cementation has increased (fig. 7-1a).

The major disconformities (Wardlaw and Henry, 1986a) and the biostratigraphic zones, based on the distribution of microfossils presented by Cronin, Brouwers, and others (1986), generally correlate readily from borehole to borehole and extend throughout the area of investigation (fig. 7-2). The sedimentary packages (SP) delimited by these disconformities (Wardlaw and Henry, 1986a) and the geologically defined material-property units (MP) proposed herein also are shown on Figure 7-2. The consistency and trends of the disconformities, the SP and MP units, and biostratigraphic zones allow reasonable prediction of pre-shot ground-zero geology for both OAK and KOA. The relationship of discontinuities, cementation zones, and general sediment type for the PEACE Program reference boreholes and the models of ground-zero geology for both OAK and KOA are shown in Figure 7-3. Excellent seismicreflection profiles (Grow and others, 1986) allow mapping of key surfaces in the undisturbed areas away from the craters, and, combined with the pre-shot geologic models, allow mapping of the probable distribution of these surfaces in a pre-shot configuration below the crater (Wardlaw and Henry, 1986b). Figure 7-4 shows the probable pre-shot surfaces at the top of the Pleistocene (disconformity 1) and at the top of the Pliocene (disconformity 5) in the KOA and OAK areas.

The most convenient way to summarize the geology for crater considerations is in material-property (MP) units. These are units delimited by major geologic horizons that best fit the material model (viz, the geologically defined units that best conform to the mechanical properties important to cratering). Differences between the sedimentary packages (SP) and materialproperty units are minor (see below) but include, for example, the pervasively cemented zone that includes SP3 and the upper part of SP4 is represented as a single unit (MP-3), although it is divided by a major disconformity (6) that represents a significant exposure surface and geologic gap.

The upper 1,200 feet of sedimentary section at Enewetak is divided into five material-property units (fig. 7-3), as follows:

- MP-1 (Holocene, Sedimentary Package 1). -- Aragonitic sediments, from the surface to disconformity 1.
- MP-2 (<u>Pleistocene, SP 2</u>). -- Aragonitic sediments with thin calcitic limestones, from disconformity 1 to 5. This unit is subdivided by disconformities 2, 3, and 4.
- MP-3 (<u>Upper Pliocene</u>, SP 3 and part of SP 4). -- Cemented interval of vuggy, calcitic limestone and aragonitic or calcitic sands, from disconformity 5 to the base of the alteration zone (see Wardlaw and Henry, 1986b, p. 25 for discussion of alteration zone). This unit is subdivided by disconformity 6.
- MP-4 (Upper Miocene-Pliocene, part of SP 4, all of SP 5). -- Aragonitic sands, from base of the alteration zone to disconformity 7. High organic content and high activity on the natural gamma logs identifies a lower subunit.
- MP-5 (<u>Miocene, SP 6, SP 7, and SP 8</u>). -- Calcitic sands and limestones, limestone variably developed, from disconformity 7 to bottom of boreholes. This unit is subdivided by disconformities 8 and 9.



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FIGURE 7-3. -- Characterization of cementation and general sediment type and relationship to material property units for reference boreholes and models for ground-zero geology. Discontinuities as lines in columns, major disconformities numbered in columns.

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FIGURE 7-4. -- Probable pre-shot surfaces for the Pleistocene (Disconformity 1) and Pliocene (Disconformity 5) in the KOA and OAK areas. Contours in feet.

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The pre-event subsurface geology in the KOA and OAK areas differs in three significant ways (fig. 7-5):

- (1). MP-2d is more consistently well-cemented in the KOA area.
- (2). MP-3 (the upper, well-cemented unit) is thicker (246 ft vs 197 ft) and shallower (top at 282 ft vs 395 ft bsl¹) in the KOA area.
- (3). MP-3 is homogeneous throughout the crater area at KOA. At OAK, this unit changes from a cemented limestone with calcitic sands beneath the reef tract to cemented limestone with aragonitic sands beneath the lagoon, and the cemented intervals appear to decrease in thickness lagoonward (contrast OAR-2/2A to OOR-17; see fig. 7-3).

In addition, the pre-event ground surfaces in OAK and KOA areas differ significantly. KOA is represented by a nearly flat shallow surface on a broad reef tract, whereas OAK is represented by a narrow, shallow reef tract, relatively steep slope, and a flat, deep lagoon bottom.

POST-EVENT GEOLOGY OF OAK AND KOA CRATERS

The excavational craters were modified profoundly by a set of processes that included shock-induced liquefaction and consolidation, subsequent flow and piping of liquefied materials from depth (both laterally and toward and/or to the surface), consequent subsidence of the region adjacent to and beneath the excavational craters, and major and repeated failures of the sidewalls of the initial and subsequent craters.

Crater Zones

OAK and KOA craters can be characterized in the subsurface by geologic, paleontologic, and seismic-reflection crater zones that, in turn, can be related to crater-event history.

Traditional crater terminology does not always adequately apply to the Enewetak craters studied; many subsurface features within the carbonate rock and sediment virtually were undescribed. Thus, limited new terminology was introduced by Wardlaw and Henry (1986b), and a few additional terms are introduced in the current Chapter (designated by an asterisk, *).

Geologic Crater Zones

 <u>Zone of sonic degradation (ZSD)</u>: the stratigraphic interval in which sonic velocities are depressed below expected velocities. Normal or (more correctly) pre-event sonic velocities are determined from the sonic signature of reference boreholes. On the multichannel-seismic profiles, the ZSD appears as a "fuzzy" area in which seismic reflectors are not coherent and are surrounded by an area where

¹ Below sea level is abbreviated bsl throughout this Volume.



MP-3: cemented limestone with calcitic sand changing to aragonitic sand lagoonward MP-3: cemented limestone with calcitic sand throughout area

FIGURE 7-5. -- Comparison of pre-shot ground surface and subsurface geology for the OAK and KOA areas.

coherent reflectors are present but downturned. The ZSD represents units of rock and sediment that are fractured or shattered, mixed, and/or otherwise disturbed significantly enough to retard the sonic velocities relative to what they were before the nuclear events occurred. All geologic crater zones lie within the ZSD (fig. 7-6).

- 2. <u>Geologic crater</u>: the subsurface expression of the crater defined by the ZSD. The geologic crater zones encountered in the central crater are as follows:
- 3. <u>Alpha 1</u> (al): <u>Mud</u>. Late-stage, fine-grained sediments with abundant brown, piped material in OAK.
- 4. <u>Alpha 2</u> (α2): <u>Graded sand</u> (distal) and <u>slumps</u> (proximal). Late-stage slope-failure and sand-turbidite flow deposits containing abundant brown, piped material. (Proximal means near material source; distal means far from the material source).
- 5. Beta la (β_{l_a}) : Graded Rubble. (*) This zone contains proximal rubble and distal sand (as in OPZ-18) with granules of rubblized material. The zone is transitional from the rubble below and slumps above and contains abundant brown, piped material near the top in the central crater area. Both Alpha 2 and Beta la show high gamma-ray activity (see fig. 7-17).
- Beta 1s* (β1_s): <u>Hiatus sand</u>. (*) Highly shocked, uppermost unit (MP-1, Holocene) sediments.
- 7. Beta 1b (β1b): Collapse rubble. (*) Thick rubble bed with sparse brown piped material within the zone in the central crater area. Both zones Beta 1a and Beta 1b are less distinct in the central-most part of the crater, and Beta 1s is missing in the same area because of mixing primarily due to late-stage piping.
- 8. Beta 2 (β2): Transition sand. Pulverized sand within the transition paleontologic zone (see below). It has a limited lateral extent. The sand grains show fractured surfaces but no internal microfracturing.
- 9. <u>Beta 3</u> (β3): <u>Rubble floatstone</u>. Rubble in which no paleontologic mixing can be shown.
- 10. Gamma (Y): Fractured and displaced rock and sediment.
- 11. Delta: (δ) Fractured but undisplaced rock and sediment.

The base of the zone of sonic degradation:

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12. Epsilon (ε): In-place, relatively unfractured stratigraphic section; outside and beneath the geologic crater.

The geologic crater zones in the debris blanket are as follows:

13. Beta la* (βla*): Graded sand and rubble *. This zone is found only in boreholes OHT-10 and OJT-12 and may be related to a large collapse and debris flow that breached the debris blanket and flowed into the

lagoon (as seen on the OAK enhanced sea-floor image, Folger and others, 1986).

- 14. <u>Beta lb*</u> (βlb*), or Beta (β) undifferentiated: <u>Rubble</u>. Debris with no brown piped material.
- 15. <u>Disturbed zone</u>: This zone represents <u>slightly altered stratigraphy</u> with no apparent discontinuities.
- 16. Delta (δ) and Epsilon (ε): Relatively unaffected stratigraphy.

The depths to various crater zones for the transition and ground-zero boreholes for both OAK and KOA craters are given in Table 7-2, and graphically displayed in fence diagrams in Figures 7-7 to 7-10. Interpretations of geologic crater zones on seismic reflection profiles through ground-zero for both OAK and KOA crater (from Wardlaw and Henry, 1986b) are shown in Figures 7-11 and 7-12.

Paleontologic Crater Zones

The paleontologic crater zones for the **central crater** follow. The depths to various paleontologic crater zones for both OAK and KOA craters are given in Table 7-3.

- 1. <u>Mixed</u>: Fossils from various biostratigraphic zones are mixed together. This zone can be crudely divided into three subzones:
 - a. Very mixed with material from mostly upper biostratigraphic zones and piped material from deeper zones.
 - b. Mixed material from most of MP-1 and MP-2 plus piped material that decreases in degree of mixing downward.
 - c. Mixed material from mostly lower biostratigraphic zones of crater and sparse piped material.

These zones were developed for KOA crater (Wardlaw and Henry, 1986b) and are applicable to OAK with minor modification. In OAK, an additional zone, represented by the "hiatus sand" (Beta 1s), occurs between paleontologic subzones b and c in the lateral part of the crater. This unit consists predominantly of Holocene (near-surface) material and shows little mixing.

2. Transition: Transitional paleontology from mixed to unmixed.

3. <u>Unmixed</u>: Paleontology in normal succession showing no mixing of materials from different biostratigraphic zones.

The paleontologic crater zones for debris blanket are:

- 4. <u>Mixed, undifferentiated</u>: generally like unit 1b within the crater, but without piped material.
- 5. Transition: as above.
- 6. Disturbed Zone: unmixed, but sparse faunas.
- 7. Unmixed: as above.

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FIGURE 7-6. -- Geologic crater model.

GEOLOGIC CRATER ZONES

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sand { } is present only in the out	ter crater of OAK.					
[PALEONTOLOGIC ZONES]	[GEOLOGIC ZONES]					
CENTRAL						
MIXED	ALPHA 1 (al) Mud					
upper biostratigraphic zones and piped material	ALPHA 2 Graded sand (distal) (α2) and slumps (proximal)					
bMixed material from most units 1 and 2 and piped material, generally decreasing in mixing downward	BETA la Graded (βl _a) rubble					
{Unmixed upper biostratigraphic zone} cMixed material from mostly lower biostratigraphic zones and sparse piped material	{BETA ls Hiatus Sand} (βl _s) BETA lb Cuilapse (βl _b) rubble					
TRANSITION	BETA 2 (β2) Transition sand					
UNMIXED	BETA 3 (β3) Rubble floatstone					
	GAMMA (Y) Fractured, displaced					
	DELTA (δ) Fractured, relatively undisplaced					
	EPSILON (ε) Relatively unfractured, in place					
DEBRIS B	LANKET					
MIXED (undifferentiated)	$\begin{array}{c} \text{BETA la} (\beta_{l_a}) & \text{Graded} \\ \text{DEBRIS} & \text{sand and rubble} \\ (\text{BETA}) & \text{Constant} \end{array}$					
TRANSITION	$\begin{array}{c} \textbf{BETA} \\ \textbf{BETA} \\ \textbf{Ib} \\ \boldsymbol{\beta} \\ \textbf{BETA} \\ \textbf{b} \\ \boldsymbol{\beta} \\ \textbf{b} \\ \textbf{c} \\$					
DISTURBED	DISTURBED					
UNMIXED	DELTA (δ) Fractured, relatively undisplaced					
	EPSILON (ε) Relatively unfractured, in place					

TABLE 7-1. -- Relationship of geologic and paleontologic zones in the crater and debris blanket modified from Wardlaw and Henry (1986b). The hiatus sand { } is present only in the outer crater of OAK.

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ZONE ¢	OBZ-4	OP Z-18	OCT-5	OTG-23	OUT-24	OKT-13	OFT-8	0IT-11
Alpha 1	198.7	201.9	163.7	-	-	164.7	130.8	155.0
Alpha 2	229.2	-	164.6	164.0	147.0	165.3	131.1	
Beta la	271.7	246.5	174.1	174.0	249.2	177.0	139.4	155.1
Beta ls	-	-	244.1	219.0	278.6	190.8	152.9	-
Beta lb	309.1	337.2	310.7	235 ?	288.0	207.0	175.4	
Beta 2	394.9	377.0	-	-	-	-	-	-
Beta 3	415.1	412.3						
Gamma	564.2	522.4	346.3	314.0	332.0	227.3	204.1	171.7
ZSD	1138.7	1082.4	863.7	842.0	830.0	831.7	639.6	697.0
ZONE ¢	0ET-7	OQT-19	0HT-10	OJT-12	ODT-6	ONT-16	ORT-20	OMT-15
 Alpha 1								
Alpha 2	106.9	-	-	-	-	-	-	-
Beta la	132.3	-	137.3	143.8	87.4	-	-	-
Beta ls	-	-	-	-	-	-	-	-
Beta 1b		-	145.2	155.0	-	135.1	-	110.9
Beta 2	-	-	-	-	-	_	-	_
Beta 3			[191.1]	[164.7]	-	[148.0]	-	[119.8]
Gamma	156.3	117.5	286.8	238.0	91.9	176.7	101.4	139.4
ZSD	505.1	413.3	587.1	387.0	311.6	242.7	239.0	223.0
		I] denot	es distu	rbed zo	ne		
ZONE ¢	OLT-14		¢	KBZ-4	кст-5	KFT-8	KDT-6	 кет-7
Alpha 1			¢	109.1				
Alpha 2	-		¢	137.3	98.9	77.8	56.2	-
Beta la	139.7		ç	167.7	120.0	96.5	79 .9	-
Beta ls	-		c	-	154.5	-	_	-

TABLE 7-2. -- Depth (ft bsl) to tops of the crater zones in OAK and KOA boreholes. ZSD = Zone of Sonic Degradation. Boreholes listed in order of increasing distance from ground-zero.

OAK AND KOA GEOLOGIC CRATER ZONES

ZONE ¢	OLT-14	¢	KBZ-4	KCT-5	KFT-8	KDT-6	KET-7
Alpha 1	-	¢	109.1	-	-	-	
Alpha 2	-	¢	137.3	98.9	77.8	56.2	-
Beta la	139.7	¢	167.7	120.0	96.5	79.9	-
Beta ls	-	¢	-	154.5	-	-	-
Beta lb	-	¢	238.5	156.1	106.0		-
Beta 2	-	¢	247.2	242.5	-	-	-
Beta 3	-	ç	266.2	259.9	-	-	-
Gamma	147.2	¢	316.2	274.3	153.8	110.1	51.1
ZSD	154.2	¢	1101.1	869.2	590.4	410.0	318.2
		¢					

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TABLE 7-3. -- Paleontologic crater zones and relation to the transition sand in OAK and KOA boreholes. Depths in ft below sea floor (ft bsf) are compatible with the footages presented in the paleontologic studies (Cronin, Brouwers, and others, 1986; Brouwers, Cronin, and Gibson, 1986; and Cronin and Gibson, 1987), which are consistently in feet below sea floor (bsf).

2233222283922		KOA C	RATER	*********	*============		
	KBZ	-4	кст-5	KFT-8	KDT-6		
Mixed Zone	0-1	37.5	0-140.1	0-28.5	0-43.6		
Transition Zo	ne 137.5-1	42 140.	1-155.2	28.5-99.3	43.6-58.5		
Transition Sa	nds 138.1-1	57.1 143.	6-161.0				
3222352333333	******			*********		# = =	
OAK CRATER							
	OBZ	-4	OPZ-18	0CT-5	окт - 13		
Mixed Zone	0-1	80	0-174	0-149	0 - 55	0-55	
Transition Zo	ne 180-2	20 17	4-211	149-187	55-68	55-68	
Transition Sa	nds 196.2-2	16.4 175.	1-210.4				
	OFT	-8	ODT-6				
Mixed Zone	0-6	4 l.	8-4.4				
Transition Zo	ne 64-7	4					
Transition Sa	nds						
		DAK CRATER	DEBRIS B	LANKET			
	OHT-10	0JT-1	2 ONT	-16 0	MT-15 OLT	-14	
Mixed Zone	0-54	0-20.	9 0-1	2.9 0	-8.9 0-7	•5	
Transition Zo	ne 54-76	20.9-67	12.9-1	4.7 8.9	-15.5	,	
Disturbed Zon	e 76-149.	5 67-94.	2 14.7-4	1.6 15.5	-28.5		
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Seismic Crater Zones

Grow, Lee, and others (1986) interpreted four subcrater seismic zones from the multichannel seismic-reflection records. They are, from top to bottom: (1) transparent zone, (2) zone of intense fracturing/depression, (3) zone of moderate fracturing/depression; and, 4, zone of minor fracturing/ depression. The zone of minor fracturing/depression has not been defined in terms of depth. The seismic zones are compared to geologic craters zones in Table 7-4.

The transparent zone corresponds to the crater fill and the transition sand (where present). In OAK, reefward of SGZ, the base of the transparent zone is difficult to interpret because some large-scale slumps (crater fill) from the reef tract are not completely transparent seismically. The bottom of the zone of intense fracturing/depression falls within gamma, the zone of fracturing and displacement in KOA, and very near the bottom of the rubble zone in OAK. The bottom of the zone of moderate fracturing/depression appears to fall close to the gamma/delta transition or that change from fractured/ displaced to fractured/in place material. The delta zone appears to be equivalent to the zone of minor fraturing/depression.

TABLE 7-4. -- Comparison of subcrater seismic zones to selected geologic crater zone boundaries for OAK and KOA craters.

SEISMIC ZONE	GEOLOGIC CRATER ZONE					
KOA						
Bottom of Transparent Zone 262 ft bsl	Bottom of Transition Sand 266.2 ft bsl					
Bottom of Zone of Intense Fracturing/Depression	Bottom of Rubble					
460 ft bs1	316.2 ft bsl					
Bottom of Zone of Moderate Fracturing/Depression	Bottom of ZSD					
755 ft bsl	1101.1 ft bs1					
OAK						
Bottom of Transparent Zone	Bottom of Transition Sand					
361 ft bs1	377.0 ft bsl					
Bottom of Zone of Intense Fracturing/Depression 590 ft bs1	Bottom of Rubble					
	564.2 ft bsl					
Bottom of Zone of Moderate Fracturing/Depression	Bottom of ZSD					
918 ft bs1	1138.7 ft bs1					

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CRATER FEATURES

Crater Material in the Lagoon

Muddy sediments in the northwestern portion of the lagoon (see fig. 7-13) are derived partly from crater material. Observations leading to this conclusion (Wardlaw and Henry, 1986b) include:

- (1). An anomalously high amount of low-Mg calcite in the sediments probably indicates mixing from diagenetically altered subsurface units.
- (2). The sediments have an anomalously high content of clay-size material, probably indicating crater-derived material. Normal lagoon sediments do not contain appreciable quantities of naturally produced clay-sized carbonate.
- (3). The sediments have measurable radioactivity, probably from the devicederived Cesium-137 (Ristvet and Tremba, 1986).

Thus, a substantial part of the mud in the northwestern portion of the lagoon (fig. 7-13) was derived from pulverization of sediment and rock particles by the nuclear detonations during the excavation of the craters. A considerable volume of fine-grained material was moved from the crater areas to the lagoon, although the volume of this lost material or proportion derived from each of the forty-one nuclear events other than OAK or KOA cannot be estimated.

Breach Deposit in the Lagoon

The enhanced sea-floor image of OAK crater displays a large flow deposit out into the lagoon (fig. 7-14). This feature extends out beyond the limits of the apparent crater, thus it, too, represents loss of material to the lagoon. This feature was not observed until after the field operations, so it was not sampled. The thickness or volume of the deposit is unknown. The deposit appears to represent a breach in the debris blanket through the "channel" (Peterson and Henny, Ch. 5 of this report, p. 5-15) and flow of material out onto the lagoon floor.



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FIGURE 7-7. -- Fence diagram of KOA boreholes showing relationship of crater Scale is vertically exaggerated 2:1. Squiggly lines represent breaks to shorten the diagram. and geologic horizons.

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FIGURE 7-13. -- Weight percent mud in bottom sediments of lagoon. Sediments in excess of 10 percent mud are probably indicative of blast-derived mud contribution.

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Piping

Brown-stained, organic-rich sediments from MP-4 were piped to the surface in substantial quantities. Several sand mounds (Halley and others, 1986) or "sand volcanoes", covered with moderate-brown, coarse-grained detritus are common on the terraces of OAK crater. This material commonly contains granule- and small-pebble-sized particles (2-64 mm) and may sparsely contain small cobble-sized materials (64-256 mm). The sand volcances observed are generally less than 10 ft high, are round to elongate, and are 16-33 ft across and up to 100 ft long. The eight volcances documented by Halley and others (1986a) are plotted on the enhanced sea-floor image (fig. 7-14). Similar features that are probably sand volcances are also shown. The volcances appear to exist in several clusters or swarms on the terraces of OAK crater. No sand volcances were observed in the KOA area; however, most surficial features have been obscured by extensive slumping and recent sedimentation (Folger and others, 1986).

Several thin sand dikes filled with brown-stained sediments, confirmed by paleontologic analysis to be from MP-4, were penetrated by the boreholes. These were inclined at a high angle to the borehole under the central crater region and terraces of OAK. Dikes were observed in boreholes OPZ-18 at 667.8 - 668.5 ft, OKT-13 at 615.0 to 615.2 ft, OTG-23 at 472.3 to 473.2 ft, and OFT-8 at 291.1 to 291.9 ft (all depths bsl; see Henry, Wardlaw, and others, 1986). No dikes were observed in the KOA boreholes.

Paleontologic Mixing

The distribution of mixed materials from different biostratigraphic zones within the geologic crater is complicated, but each fossil is a clue to unraveling the history of formation of crater-fill deposits. In addition to the general three to four mixed zones presented in the previous section, both KOA and OAK have an overprint of hydraulic sorting in the central region due to post-deposition upward flow of piped material from strata below the excavational crater. In KBZ-4, the piped material shows hydraulic sorting of various fossil groups (see Brouwers, Cronin, and Gibson, 1986). In OBZ-4 and OPZ-18, the faunas are depleted and represented by sparse piped material in the lower part of the crater fill (fig. 7-15; and Cronin and Gibson, 1987), thought to indicate preferential removal of contained faunas by hydraulic flow and scant deposition of MP-4 faunas.

The mixing within the crater is displayed in Figure 7-16 for OAK and Figure 7-17 for KOA The biostratigraphic zones represented are defined in the reference boreholes in sequence of superposition and with increasing depth are: surficial (S), AA, BB, CC, DD, EE, FF, and GG. Piped material from depth designated as "piped" in the figures is represented by biostratigraphic zones II, JJ, KK, LL, and MM. Because the KOA event excavated down to the DD/EE zone boundary, most EE and all FF material in the crater-fill indicates shallow piping. Because the OAK event excavated down to a point within EE, possibly some EE and all FF and GG material in the crater-fill represents shallow piping. Each crater will be briefly discussed from bottom up (or as they filled).



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FIGURE 7-14. -- Distribution of observed (solid circles) and probable (dashed circles) sand volcanoes shown in clusters on enhanced sea-floor image of OAK crater and location of breach and flow deposit in lagoon.

DEEP-PIPED MATERIAL OBZ-4 **OPZ-18** % 10 0 20 0 10 0 15 200 ec.l ex2 Bla Ala 300 -BIL 611 400 FEET

FIGURE 7-15. -- Number of specimens (#) from MP-4 and MP-5 (minor) and percent (%) of total ostracoles picked in crater zones in boreholes OBZ-4 and OPZ-18.



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biostratigraphic zone.

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OAK Crater.

The lower mixed subzone (C, figs. 7-15 and 7-16) and the transition zone coincide for much of OAK crater (this is because of the limits of resolution, similarity of taxa in the zones, and gradational nature of these zones). This zone contains undifferentiated EE-FF material with sparse AA and deep-piped material in the central crater and of slightly mixed material from progressively stratigraphically higher zones outward to OFT-8 where it contains mostly BB-CC material.

The middle mixed subzone (B) represents a maximum of mixing of material in the central-crater area with components from zones AA-GG mixed with abundant deep-piped material in the upper part of Beta 1b and the lower part of Beta 1a. Laterally, this interval is represented by the "hiatus" sand (Beta 1s) which consists largely of AA material on top of Beta 1b. This situation is complicated at OCT-5 by an apparent local slump that covers and possibly involves the "hiatus" sand (fig. 7-16).

The <u>upper mixed subzone</u> (A) can be divided into two parts in the centralcrater area and the inner terraces (OCT-5). In the central-crater area, the lower part consists of common AA-CC material and deep-piped and sparse DD-GG material, and the upper part consists of abundant surficial and AA components with common BB-GG and deep-piped material. Under the inner terraces (OCT-5), the upper mixed subzone consists of a lower part with abundant AA and common BB-GG material (no deep-piped material) and a upper part that is highly mixed with AA-GG and deep-piped material. Laterally, the upper mixed subzone commonly consists of very mixed AA-GG material decreasing outward to AA-FF material with sparse deep-piped material at its top. At OKT-13, the base of the upper mixed subzone (which coincides with the base of Beta 1a) is mixed with material from the underlying unit, the "hiatus" sand (Beta 1s).

Deep-piped material from MP-4 is mixed throughout the crater-fill in both OBZ-4 and OPZ-18, suggesting that the central crater bowl served as the common avenue for venting of MP-4 material. Although sand volcanoes are common on the terraces, mixing of deep-piped material from MP-4 is restricted to surface or near-surface deposits, suggesting that the volcanoes are a late-stage feature and did not represent the more common avenue of venting. Venting under the terraces probably did not take place until significant concentric fracture zones opened sufficiently in the subsiding crater to serve as conduits.

<u>Shallow-piped material</u> is that material in the crater-fill from shallow biostratigraphic zones that remained completely below the excavational crater. In OAK, this material is represented by components of biostratigraphic zones FF-GG. Shallow-piped material is common in the upper mixed subzone throughout the crater-fill, common in the middle mixed subzone, and sparse in the lower mixed subzone in the central crater.

KOA Crater.

The <u>lower mixed subzone</u> (C, fig. 7-17) and the <u>transition zone</u> coincide in most of KOA crater (for the same reasons as in OAK) and consist of CC-DD material with sparse BB and EE-GG material at KBZ-4, of DD-EE material at KCT-5, and of BB-CC material at KFT-8 and KDT-6.

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The middle mixed subzone (B) consists of AA-EE material throughout the crater-fill. In addition, at KBZ-4, this zone contains deep-piped material that indicates hydraulic sorting in Beta la with FF-II ostracodes and KK-LL foraminifers, and a normal distribution of deep-piped FF-MM ostracodes and foraminifers in Alpha 2. A very thin, muddy "hiatus" sand may be preserved in KCT-5 within the middle of this unit.

The upper mixed subzone (A) is confined to Alpha zones. At KBZ-4, it is dominated by surficial (S) and AA-CC material with sparse DD-GG and deep-piped (KK-LL) material. At KCT-5, it consists of microfossils from AA-EE with very sparse deep-piped material. At KFT-8, the upper mixed subzone cannot be differentiated from the middle mixed subzone, and the whole interval consists of AA-EE material. At KDT-6, it is dominated by AA-BB with CC-EE and deeppiped (HH-LL) material.

Deep-piped material from MP-4 is mixed with other material throughout most of the crater-fill in KBZ-4. It is only found in surficial deposits in the transition boreholes. This suggests that the central bowl in KOA, which is now obscured by pervasive slumping, served as the common avenue for venting deep material from MP-4 just like in OAK.

In KOA, <u>shallow-piped material</u> is represented by components of biostratigraphic zones EE-GG, predominantly EE. The KOA crater-fill material shows much more pervasive shallow piping than in OAK. This shallow piping obscures some of the mixing subzones and yields fairly common mixed faunas of AA-EE. The pervasiveness of the mixing also implies that shallow piping occurred over a broad area. In addition, the paucity of samples and boreholes and a less rigorous study of the KOA material gives less definition of the mixing in KOA.

Estimates of Volume of Piped Material.

The volume of deep-piped material can be estimated with the techniques developed for the detailed paleontologic analysis of the OAK crater by Cronin and Gibson (Ch. 3 of this Report). Deep-piped material occurs only near the surface outside the central bowl and is essentially negligible in quantity. If all grain sizes behaved as those between 63 through 850 μ (the size range from which ostracodes are extracted) and if sedimentary particles of different shapes and densities (minor, all CaCO3) behave the same as ostracode valves and carapaces, then the detailed percentages of piped ostracodes reflect the entire sedimentary assemblage (Cronin and Gibson, Ch. 3 of this report). A conservative volume estimate based on these data is 4.83 million cubic feet (5.1 % of the total volume of central bowl to a depth of 149 ft with a radius of 450 ft from GZ).

A semiquantitative approach also can be attempted for estimating the shallow-piped material in OAK. Shallow-piped material is identified as those ostracodes that characterize the FF/GG zones, those zones that remained completely below the excavational crater. Shallow-piped material is similar in distribution to deep-piped material within the central bowl, it occurs throughout the crater fill. Because of the general low abundance of FF/GG zone indicators, any patterns in the distribution within the crater-fill is difficult to discern. The crater bowl probably was an avenue for shallow piping, and the piping probably obscures any patterns of distribution in a manner similar to that for deep-piped material. Ostracodes that characterize the FF/GG zones are typically sparse, averaging 0.4 percent in the faunas above the FF/GG zones in the reference boreholes. They average a sparing 7.5 percent in the faunas of the zones that they characterize in the reference boreholes. These ostracodes average 3.3 percent in the central crater-fill faunas. This implies a whopping 41 percent of the central crater-fill material may have been derived from the FF/GG zones. A volume estimate based on these data is 45.62 million cubic feet of shallow-piped material within the central bowl. However, unlike deep-piped material, shallow-piped material is distributed in significant quantities in Beta 1a and Alpha zones outside the central bowl was piped.

Paleontologic Model of Crater-Fill.

The paleontologic zonation of the crater-fill can be summarized into a simple model that is applicable to both craters studied. It is extremely relevant for constraints on timing of processes of crater-filling. It is presented in Figure 7-18. The zone of shallow-piped material coincides with that of the deep-piped material through mixed subzones B and C in the central crater but encompasses all of mixed subzone A throughout the crater. The zones of piped material indicate the relative timing of arrival of material to the surface. Shallow-piped material first arrived to the surface after the deposition of the hiatus sand (Beta 1s), which probably resulted from washback. Deep-piped material first arrived to the surface after the deposition of the graded rubble (Beta 1a), during deposition of Alpha.

The zone of piped material from depth has a strong overprint of mixing and hydraulic sorting in the central bowl, especially in mixed subzone B, where abundant deep-piped material was deposited. The central bowl served as the probable avenue for venting of the deep-piped material. Shallow-piped material also appears to have vented, in part, through the central region. However, shallow piping appears to have occurred throughout the crater wings which implies venting throughout the crater region.

Injection

Holocene sediment (from MP-1) appears at an anomalous depth in borehole OPZ-18 within the transition sand (390.6 to 410.0 ft bsl) and in thin dikes below the transition sand (434.5 to 435.2 ft bsl, and questionably at 415 ft). This appears to be injection of near-surface material at the base of the excavational crater.

Gamma Activity

In Enewetak boreholes, elevated gamma activity appears to reflect the following: (1) the presence of device-produced radionuclides; (2) the presence of brown-stained, organic-rich sediments from MP-4; and (3) various other factors. For example, a gamma peak of the third type occurs within muddy sediments overlying a discontinuity in OIT-11 (fig. 7-19). It appears that other peaks of the third type also can be related to thick zones of "teabrown" (organically stained) micrite cement.

PALEONTOLOGIC MIXED ZONES

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- A MOSTLY UPPER BIOSTRATIGRAPHIC ZONES
- B VERY MIXED
- C MOSTLY LOWER BIOSTRATIGRAPHIC ZONES

FIGURE 7-18. -- Paleontologic model of the crater-fill and the paleontologic mixed zones.





Cesium-137 levels coincide directly to the gamma activity in the Alpha, Beta-1, and Beta-2 crater zones (Ristvet and Tremba, 1986). Furthermore, the gamma activity of the brown-stained, organic-rich sediments is caused largely by naturally occurring isotopes of thorium and uranium. These radionuclides were not observed within the Alpha, Beta-1, and Beta-2 crater zones.

Naturally occurring thorium and uranium isotopes were detected in borehole OIT-11 and probably account for the "other" peak (third type) in gamma activity noted above. Similarly, in borehole OHT-10, a small peak in the gamma activity probably reflects naturally occurring thorium and uranium (fig. 7-19).

Figures 7-19 and 7-20 compare the gamma log, paleontologic and geologic crater zones, and general lithologies for KOA and OAK crater areas. Only boreholes on transects with full geologic sampling and open-hole gamma logs were utilized for this comparison. Essentially, the gamma logs confirm the general trends in radionuclide abundance (Ristvet and Tremba, 1986; and fig. 7-21). The Beta-2 / Beta-3 boundary (where present) and the Beta-1 / Gamma boundary (where present) appear to represent the demarcation between occurrence and absence of device-produced radionuclides. Naturally occurring radionuclides appear to reflect the presence of deep-piped material in the Beta-3 crater zone. Device-produced radionuclides are most abundant within the bottom of Alpha-2 (graded sands) and top of Beta-la (graded rubble) in OBZ-4 and OPZ-18, respectively. In KBZ-4, they are most abundant at the base of Alpha-1 and at the top of Alpha-2. In KCT-5, there is only a trace of radionuclides (device-produced and natural). In borehole OCT-5, deviceproduced radionuclides are most abundant within the lower part of the Beta-la crater zone. In OKT-13, device-produced radionuclides show two peaks, one within Beta-la and the other (larger) near the base of Beta 1b.

Distribution of Radionuclides

The distribution of radionuclides within OAK crater is shown in Figure 7-21 (Ristvet and Tremba, 1986). In OBZ-4, the device- produced radionuclide (Cesium-137) is common in Alpha 1, Alpha 2, and Beta 1a, with peak abundance in Alpha 2. Most of the crater-fill in OPZ-18 consists of muddier sediments than OBZ-4 and consequently contains higher concentrations of Cesium-137. In OPZ-18, radionuclides are common to Alpha 1, Beta 1a, Beta 1b, and Beta 2; the Beta 2 occurrences represent the injected material. Peak abundance is in the upper part of Beta 1a. A moderate amount of cesium is found in OKT-13 below and above the "hiatus" sand (Beta 1s), in Beta 1b, and in Beta 1a and Alpha, respectively.

Radionuclides are sparse in KOA crater and only common within KBZ-4. Here, they mimic the gamma-ray profile (fig. 7-20), with peak abundance in Alpha 1 and a trace at the base of Beta 1b and top of Beta 2.

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crater zones, and gamma-ray logs for selected boreholes, KOA crater, and index maps for the KOA and OAK craters. Symbols the same as Figure 7-19. FIGURE 7-20. -- Borehole lithology (L), geologic (G) and paleontologic (P)

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Distribution of Shocked Calcite

The distribution of shocked calcite is shown in Figure 7-21 (Polansky and Ahrens, Ch. 4 of this report). Only sparse, possibly moderately shocked calcite is present in the central crater area within Beta 1b. Highly shocked calcite is found within the injected material in OPZ-18. Under the terraces, possibly moderately shocked calcite is found in Beta 1a, Beta 1b, and Gamma with highly shocked calcite in the "hiatus" sand, Beta 1s. Outside the limits of crater-derived rubble (Beta zone), highly shocked calcite occurs in Alpha 2 and Beta in OET-7 and in Alpha 2 in OAR-2A. The material in OAR-2A represents post-event deposition of shocked calcite away from the crater, probably by the sweeping away of fine-grained ejecta from the reef tract by currents and redeposition of it in the area of OAR-2A. The material in OET-7 in graded sands (Alpha 2) seems to represent post-event deposition like that in OAR-2A. The material in OET-7 in the undifferentiated rubble apparently represents buried ejecta.

Depression and Uplift of Structural Surfaces

The surface at the top of the Pleistocene in both the OAK and KOA areas shows a pattern of central removal and lateral depression on the net-change (delta) figures (figs. 7-22A and 7-23A, respectively) derived from the preand post-shot surface contour maps (figs. 7-4 and 7-24). In addition, in the OAK area, two lateral depression troughs are developed along the pre-shot slope from reef to lagoon. Also, the Pleistocene surface appears to be irregularly disrupted or preserved beneath the debris blanket (dashed lines, fig. 7-22A) and irregularly uplifted near the margins of the debris blanket. The maximum current depression observed in OAK is 63 ft and in KOA is 53 ft. The maximum uplift in OAK is 14 ft. No strata in KOA are currently uplifted.

The surface at the top of the Pliocene in the OAK area (fig. 7-22B) shows central concentric depression slightly skewed toward the reef and a broad region of shallow uplift beneath both the debris blanket and the lagoon. The maximum depression is 193 ft beneath GZ. The maximum uplift appears to be about 21 ft. The Pliocene surface in the KOA area certainly was influenced by detonation of the MIKE device (fig. 7-23B). KOA shows a complicated pattern of depression with maximum depression on the lateral wings away from GZ. The pattern of depression from MIKE crater area would suggest that the area in the proximity of KOA GZ experienced 0 to 10 ft depression and the entire region from KOA GZ to MIKE experienced progressively greater depression toward MIKE. This possibly influenced the apparent lateral extension in depression roughly perpendicular to the line from KOA ground-zero to MIKE ground-zero.

COMPARISON OF OAK AND KOA CRATERS

The following comparisons and contrasts can be made between KOA and OAK craters:

(1). The base of the zone of sonic degradation (ZSD) is similar in both craters -- 1,139 ft bsl for OAK ground-zero (GZ) and 1,101 ft bsl for KOA GZ. The ZSD appears to form a narrower cone at KOA

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FIGURE 7-22. -- Maximum depression/uplift of Pleistocene and Pliocene surfaces, OAK crater. Pleistocene surface is projected beneath debris blanket, where it is disrupted but probably remains as several isolated outliers such as encountered in OIT-11. Surface is lightly stippled where removed, heavily stippled where uplifted.



FIGURE 7-23. -- Maximum depression of Pleistocene and Pliocene surfaces, KOA crater. Surface is lightly stippled where removed.

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FIGURE 7-24. -- Present-day (post-shot) location of Pleistocene and Pliocene surfaces, KOA and OAK craters. Contours in ft below H&N datum (bsl).

- (2). MP-3, the significantly cemented and altered zone, was probably 246 ft thick with the top at 280 ft bsl at KOA GZ and 183 ft thick with the top at 400 ft bsl at OAK GZ.
- (3). MP-3 is depressed approximately 193 ft at OAK GZ and approximately 89 ft at KOA GZ. MP-3 shows depression and fracture at OAK and, in addition, at KOA, shows apparent rebound in the central part of the crater.
- (4). At ground-zero, the Alpha zone (mud and graded sand) is comparable between KOA and OAK; however, the Beta zone (rubble) is twice as thick at OAK. In particular, Beta 3 (rubble floatstone) is much thinner at KOA. The total lateral extent of the Beta zone is nearly the same at both craters.
- (5). The transition sand (Beta 2) is more extensive in KOA than OAK, with an average diameter of approximately 918 ft at KOA and 816 ft at OAK. The transition sand is more elongate oval at KOA than at OAK.
- (6). The collapse rubble (Beta 1b) is similar in both craters, although thicker in OAK. The Beta 1b zone thins toward the lagoon at OAK and thins toward MIKE crater at KOA.
- (7). The hiatus sand (Beta 1s) is much less extensive at KOA, presumably due to thinner and shorter-term deposition and to more extensive destruction by late-stage collapse.
- (8). The graded rubble (Beta la) is similar in both craters. The rubble becomes thicker and muddier in the direction of the lagoon at OAK (i.e., toward its distal margin) and in the direction of MIKE crater in KOA For all intent and purpose, for the KOA event, MIKE served as a "lagoon" similar to the natural lagoon off OAK, but much smaller in extent.
- (9). The graded sands (Alpha 2) are similar in both craters. This zone is common throughout the KOA crater but absent near the bathymetric center (OPZ-18) of OAK crater.
- (10). Mud (Alpha 1) occupies the central region of both craters.
- (11). A debris blanket is extensive on the lagoon side of OAK; only two possible debris mounds of limited distribution exist on the MIKE-side of KOA.
- (12). Deep-piped material is common only in Alpha 2 in KOA and probably vented in a limited area at the central crater. Deep-piped material is common to Alpha 1, Alpha 2, and Beta 1a in OAK and probably vented in an extensive area of the central crater and terraces.
- (13). Shallow-piped material is common to Alpha and Beta la zones throughout the crater wings and found in all zones in the central crater in both craters. In KOA, it is represented by EE-GG material in the central

crater and EE material in the crater wings. In OAK, it is represented by FF/GG material throughout the crater.

- (14). KOA crater is characterized by late-time sedimentation exceeding subsidence. OAK crater, in contrast, is characterized by late-time subsidence exceeding sedimentation.
- (15). Device-produced radionuclides appear to be mostly limited to the Beta 2 and overlying zones in the craters. Radionuclides were detectable only in KBZ-4 for the KOA crater. In OAK crater, peak abundance of device-produced radionuclides progressively moves down in the crater zones away from GZ. For example, the peak abundance is in Alpha 2 in OBZ-4, at the top of Beta 1a in OPZ-18, at the bottom of Beta 1a at OCT-5, and at the bottom of Beta 1b at OKT-13.

GEOLOGIC CRATER MODEL FOR OAK AND KOA

The transition sand (Beta 2) represents the remnants of the base of the excavational crater. It is characterized by sand-sized material that is formed by fracture and pulverization, by its transitional nature from mixed paleontology to unmixed paleontology within it, and by its containing injection dikes and debris. That shocked calcite is not common within the transition sand is due to two factors: (1) the sampling technique used in which granule- and larger-sized clasts were predominantly sampled (Polansky and Ahrens, 1987, Ch. 4 of this report), and (2) the relatively low shock pressures that probably existed in this region at formation (< 15 kilobars).

The rubble floatstone (Beta 3), beneath the base of the excavational crater (Beta 2), may represent fracture and disruption of sediment and rock caused by the maximum growth of the transient crater.

The collapse rubble (Beta 1b) represents crater-sidewall and partial flap collapse. This zone reflects paleontologic mixing of zones near the base of the excavational crater. The asymmetric crater at OAK demonstrates partial sidewall and flap collapse and movement down the resulting slope away from the transient crater to form the majority of the debris blanket. The part of the flap involved in craterward collapse is that closest to the sidewall which would represent paleontologic zones contained in the sidewall itself. That the paleontologic mixing seems to reflect mixing of material from zones near the base of the crater suggests that most of this unit was deposited rapidly as a single, major, crater-wide collapse feature. This major collapse appears to have destroyed the lateral part of the excavational crater base and its sidewalls. The highly mixed material in the central crater bowl represents a variety of depositional modes that may include wash-back and piping of sufficient magnitude to keep the central bowl "boiling" (continuously mixing). This part of the unit was deposited contemporaneously with the "hiatus" sand (Beta 1s) which represents wash-back and a brief period of quasi-stabilization of the crater and deposition of post-event sediments. The "hiatus" sand is well sorted and contains the highest concentration of shocked calcite indicating deposition from wash-back and fall-back, but, curiously, contains no radionuclides. At OAK and KOA craters, the occurrences of radionuclides are spotty. The decades that have transpired since the event have allowed many radionuclides to dissipate (Ristvet and Tremba, 1986). The

remaining commonly detectable radionuclide is Cesium-137. It is associated with muddy sediments (McMurtry and others, 1985; Wardlaw and Henry, 1986b) and may have been preferentially deposited with muds, and, therefore, would not be common in well-sorted sands. Cesium-137 is involved in progressively younger and muddier deposits in the crater-fill toward ground-zero. Its absence in the "hiatus" sand probably indicates the winnowing out of silt and finer grains during the wash-back / fall-back process.

The graded rubble (Beta 1a) represents deposition probably caused by several major slumps. This indicates that subsidence significantly destabilized the existing crater margins and resulted in collapse. One such collapse in OAK appears to have originated on the reef side. Material from this collapse flowed through the crater and breached the debris blanket, leaving deposits on top of the debris blanket (OHT-10, OJT-12), and flowed out into the lagoon, as seen in the enhanced sea-floor image of OAK (fig. 7-14).

The graded sands and slumps (Alpha 2) represent late-stage, local collapse and deposition of the expanding and subsiding crater margins.

Late-stage mud (Alpha 1) represents post-event, low-energy deposition within the central crater. The differences in Alpha 1 and distal Alpha 2 sands are slight, as shown by the sediment analysis by Melzer and Patti (written communication, 1987).

The idealized distribution of these crater units is shown for a symmetric crater (KOA, fig. 7-25A) and for a asymmetric crater (OAK, fig. 7-25B). The gradational units beneath the transition sand that represent gradually less-stressed sediment and rock within the significantly fractured zone of sonic degradation are also shown.

Thinning Analysis

This analysis simply compares the pre-shot model of inferred horizon location to the measured post-shot position. The comparison of positions is shown in Figures 7-26 to 7-29 and Tables 7-5 and 7-7. The analysis is displayed graphically in Figures 7-30 and 7-31 and tabulated in Tables 7-6 and 7-8. The upper correlation line in Figures 7-30 and 7-31 correlates the preshot model to the probable original stratigraphic depth now preserved beneath crater-fill (where present).

Stratigraphic Density Profile

The analysis of the borehole gravity surveys (Beyer, Ristvet, and Oberste-Lehn, 1986; and Beyer, Ch. 2 of this report) provide valuable information about bulk density of the strata in the vicinity of OAK crater. By averaging the borehole-gravimetry results within stratigraphic units, the density change can be compared directly with inferred stratigraphic thinning or thickening for areas where borehole gravity surveys were taken. Figure 7-32 and Table 7-9 relate the gravimetry results to the stratigraphic units. MP-2a appears anomalously dense in the reference sections. The average of MP units 2a-c in the reference sections is utilized to compensate for this anomalous density, especially to compare to crater-fill material in the analysis.

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FIGURE 7-25. -- Idealized model of geologic crater for a symmetric crater (A) and an asymmetric crater developed on a significant slope (B).

UNIT	OB2	2-4	OPZ-18		0CT	-5 	0T0	G-23	TUO	-24	
1	14	-	47	-	16.5	-	45	-	2	-	
2a	123	-	128	-	115	-	128	-	106	-	
2Ъ	165	-	166	-	155	-	168	-	145	-	
2c	265	-	2/5	-	255	368.4	253	· · · · ·	225	3/3.0	
20	315	502 0	410	502.0	305	41/•9	327	434.0	2/0	407.0	
38 35	544	701.2	555	723.5	534	4J2•7 572.2	544	610.0	490	592.0	
42	592	747.3	600	761.9	585	623.7	594	669.0	528	630.0	
4b	775	847.7	765	809.9	785	799.7	766	787.0	784	784.0	
5a	950	1013.8	956	1000.0	945	944.6	1000	1000.4	925	925.0	
5b	1050	1065.1	1050	1063.0		-	-	-	1025	1025.0	
5c	1115	1114.6	1114	1114.0	-	-	-	-	-	-	
UNIT	061	[=13 	OFT-8			0IT-11		OET-7		0QT-19	
1	102	-	16	-	122	-	18	-	46	-	
2a	141	-	115	204.1	147	185.4	118	173.4	129	168.0	
2ь	170	232.9	155	223.3	20 9	247.4	167	220.6	195	233.9	
2c	275	326.6	230	272.0	-	-	279	294.7	240	274.7	
2d	362	411.6	305	344.6	345	375.0	305	320.4	330	365.3	
3a	410	431.3	390	419.8	405	434.8	395	410.0	406	413.3	
30 (-	547	504.0	535	202.0	545	502.0	540	555.0	548	548.5	
4a 4b	290 766	745 0	70%	70/ 0	750	759 0	292	702 0	200 767	20/•2 766 5	
40 50	975	974 5	025	925.0	080	080 0	025	025 D	1020	1020 1	
ጋa 5 እ	1037	1036.5	92 J -	92 J • 0	900 -	-	92J -	929.0	1020	-	
5c	-	-	-	-	-	-	-	-	-	-	
UNIT	онт	-10	017	5-12	ODT-6		ONT-16		ORT-20		
1	124	-	115	-	20	-	132	-	70	-	
2a	152	-	149	-	116	161.5	149	-	130.5	160.6	
2Ъ	212	213.3	216	238.0	160	201.3	219	238.6	187	216.2	
2c	-	-	-	-	231	231.3	-	-	243	262.7	
2d	361	360.8	350	350.0	315	315.0	338	337.8	327	346.7	
3a 25	419	403.4	405	390.3	397	397.0	407	395.2	405	411.7	
jD Ka	54/	551+4	54/	531.0	546	546.0	550	537.9	552	552.0	
4a 46	000	284+U	000	204.U	294 202	294•U	715		200	200.4	
40 5a	121	/21+1	001	132.0	192	192.0	/10	/10.0	101/	/0/•U	
Ja 5h	70/ -	70/•3	991 -	-	725	923+0	774 _	773.0	- 1014	1012•2	
5c	-	-	-	-	-	-	-	-	-	-	

TABLE 7-5. -- Depth (ft bsl) to MP unit boundaries, pre- and post-shot, OAK crater. Boreholes listed in order of increasing distance from ground zero.

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(TABLE 7-5 continued on next page.)

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UNIT	OM	F-1 5	0LT-14		
1	142	-	132	-	
2a	153	140.8	158	159.4	
2b	225	225.0	227	227.0	
2c	-	-	-	-	
2 d	335	334.6	341	341.1	
3a	395	373.9	399	383.8	
3Ъ	551	529.7	550	534.6	
4a	600	579.0	600	585.0	
4b	702	701.9	700	700.0	
5a	1014	1013.5	1010	1010.2	

TABLE 7-5. (continued from preceeding page.)

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Distance of boreholes from surface ground-zero, in feet:

OBZ-4 OPZ-18 OCT-5 OTG-23 OUT-24 OKT-13 OFT-8 OIT-11 OET-7 OQT-19 OHT-10	7 335 658 804 858 989 1129 1206 1375 1444 1462	
OFT 9	1120	
01-8	1129	
01T-11	1206	
OET-7	1375	
0QT-19	1444	
OHT-10	1462	
0JT-12	1696	
ODT-6	1715	
ONT-16	1827	
ORT-20	1846	
OMT-15	2204	
OLT-14	2754	
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UNIT	OBZ-4	OPZ-18	0CT-5	OTG-23	OUT-24	OKT-13
1* 1 2* 2a 2b 2c 2d 3a 3b 4a 4b 5a 5b	$ \begin{array}{cccccccccccccccccccccccccccccccccccc$	100 - 17 - 08 - 49 10 11 15 71 33 00 33 28 20	100 - 40 - - 01 69 05 04 00 12 11 09 00 00	100 - 26 - 19 - 39 00 00 00 31 18 09 00 00 00	100 - 28 - - 30 33 37 00 00 00 00 00 00 00 00	100 - 11 - 11 15 02 59 03 02 01 10 05 00 00 00
UNIT	OFT-8	0IT-11	0ET-7	0QT-19	ODT-6	ORT-20
1* 1 2* 2a 2b 2c 2d 3a 3b 4a 4b 5a 5b	100 - 00 52 35 22 03 12 00 00 00 00 00 00 00 00 00 0	40 09 00 00 06 03 09 07 00 10 04 00 00 00	14 37 00 04 34 15 01 00 00 00 00 00 00 00 00 00 00 00 00	23 21 00 09 09 11 00 37 05 04 00 00 00 00 00	10 14 00 10 68 16 00 00 00 00 00 00 00 00 00 00 00 00	00 02 00 17 09 00 17 09 00 17 09 05 04 00 00 00 00 00
UNIT	онт-10	0J T- 12	ONT-16	OMT-15	OLT-14	
1* 1 2* 2a 2b 2c 2d 3a 3b 4a 4b 5a 5b	$ \begin{array}{c} 100 \\ - \\ 09 \\ - \\ 13 \\ 27 \\ 00 \\ 00 \\ + 11 \\ 00 \\ 00 \\ 00 \\ 00 \end{array} $	$ \begin{array}{c} 100 \\ - \\ 14 \\ - \\ 16 \\ 18 \\ 27 \\ 00 \\ 00 \\ + 12 \\ 00 \\ 00 \\ 00 \\ 00 \\ 00 \\ 00 \\ 00 \end{array} $	$ \begin{array}{c} 100 \\ - \\ 02 \\ \hline 17 \\ 17 \\ 00 \\ 00 \\ +10 \\ 00 \\ 00 \\ 00 \\ 00 \\ 00 \\ 00 \end{array} $	$ \begin{array}{c c} 89 \\ - \\ 00 \\ +17 \\ 00 \\ 10 \\ 35 \\ 00 \\ 00 \\ +21 \\ 00 \\ 00 \\ 00 \\ 00 \\ 00 \\ 00 \\ 00 \\ 0$	$ \begin{array}{c} 53 \\ - \\ 00 \\ 02 \\ \hline 00 \\ 07 \\ 18 \\ 00 \\ 00 \\ +15 \\ 00 \\ 00 \\ 00 \\ 00 \\ 00 \\ 00 \\ 00 \\ 0$	

TABLE 7-6. -- Thinning/thickening analysis (in percent) of MP units beneath OAK crater. * = percent removed, where removal has occurred lower number indicates thinning in remaining sediments; + = percent thickened; averages indicated by brackets are weighted.

TABLE 7-7. -- Depths (ft bsl) to MP unit boundaries, pre- and post-shot, KOA crater. Boreholes listed in order of increasing distance from ground-zero.

UNIT	K	8Z-4	K	CT-5	KF	'T-8	K1	DT-6	K	ET-7
1	+7	-	7	-	6	-	5	-	5	-
2a	37	-	35	-	33	-	33	-	32	-
2Ъ	145	(247.2)	146	(242.5)	146	233.1	147	166.8	148	148.0
2c	193	287.0	195	274.3	195	257•4	195	215.0	195	195.0
2d	250	344.0	246	365.4	244	372.0	243	282.0	242	242.0
3a	282	368.6	285	392.9	285	395.6	286	382.0	288	368.0
3Ъ	470	480.7	470	-	470	-	470	-	470	-
4a	528	539.0	526	593.0	525	590.0	525	581.0	525	575.0
4b	820	848.1	820	820.0	820	820.0	820	820.0	820	820.0
5a	979	979.0	996	996.0	999	999.0	1005	1005.0	1008	1008.0
5b	1090	1089.6	-	-	-	-	-	-	-	-
5c	1147	1147.3	-	-	-	-	-	-	-	~

Distance of boreholes from ground-zero, in feet: KBZ-4 12 KCT-5 645 KFT-8 870 KDT-6 1182 KET-7 1326

TABLE 7-8. -- Thinning/thickening analysis (in percent) of MP units beneath KOA crater, symbols as in Table 7-6.

UNIT	KBZ-4	KCT-5	KFT-8	KDT-6	KET-7	
1*	100	100	100	100	00	
1	-	-	-	-	+110	
2*	44	55	17	00	00	
2a	- T	- T	- T	49	50	
2b	18 12	+64 +08	49 +16	00 +07	00 +08	
2c	00	+78	+133	+39	00	
2 d	24	30	57	+132	+173	
3a	41 31	T 17	<u>⊤</u> 19	17	13	
3b	00	1]	
4a	+ <u>05</u> 03	23 15	23 14	19 12	17 11	
45	18	00	00	00	00	
5a	00	00	00	00	00	
5Ь	00	00	00	00	00	



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FIGURE 7-31. -- Graphic thinning analysis for KOA crater boreholes as in Figure 7-30.

	AVERAGE DENSITIES FOR STRATIGRAPHIC MATERIAL PROPERTY UNITS AND CRATER ZONES											
	MODEL	ORT-20	0QT-1 9	OTG-23	0PZ-18							
			~~~~~~~~~~~~~~~~~~~~~~~~~~~~~~~~~~~~~~~		* <b>*</b> ***************							
					1.812(06*)	α						
				Í	1.876(02*)	β1,						
1		1.855	1.895		1.982(+.03*)	βı						
2a	1.965	1.923(02)	1.921(02)		1.967(+.03*)	β2້						
2Ъ	1.908	1.918(+.01)	1.904( - )		2.027(+.06*)	β3						
2c	1.919	1.923( - )	1.942(+.01)	2.017(+.05)	2.124(+.11)	2c						
2d	1.920	1.985(+.03)	1.982(+.03)	2.030(+.06)	2.167(+.13)	2d						
3a	1.978	1.985( - )	2.018(+.02)	2.052(+.04)	2.116(+.07)	3a						
3Ъ	1.976		1.950(01)	2.063(+.04)	2.080(+.05)	3Ъ						
4a	2.011		2.025(+.01)	2.150(+.07)	2.045(+.02)	4a						
4b			1.959		1.993(+.02)	4b						

TABLE 7-9. -- Stratigraphic bulk density analysis. Values in gm/cc. Change from bulk density model (the composite reference sections) is indicated in parenthesis.

The model is the average value for units in the reference boreholes OOR-17 and OSR-21, asterisk (*) indicates the difference in bulk density from average value value of normal sediments (2a-c) which is 1.919, OQT-19 is used for model density value of unit 4b.

> $\alpha$  = Alpha  $\beta_{1a}$  = Beta la  $\beta_{1b}$  = Beta lb  $\beta_{2}$  = Beta 2  $\beta_{3a}$  = Beta 3a









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# General Densification and Flow Patterns

# OAK Crater.

The density values measured for the crater zones Alpha and Beta in OAK (OPZ-18) are compared to the average value of normal (undisturbed) sediments in MP-2(a-c). The author feels that the average value for MP-2a through MP-2c also adequately characterizes the upper sediments that were not measured by gamma-gamma density or borehole gravimetry (MP-1 and part of MP-2a). Material in the upper crater zones (Alpha and Beta 1a) in OAK appear to be less dense than normal sediments. Beta 1b and Beta 2 appear to be slightly more dense than normal sediments. Beta 3 has significant densification. Beneath the rubble zone (Beta), the rock and sediment occur in normal stratigraphic order, and density and thinning can be compared directly to the reference sections.

MP-2c (OPZ-18 and OTG-23) immediately subjacent to the Beta zone is moderately densified. Thinning of this unit cannot be calculated because the top of the unit is not preserved in either borehole. However, the significant densification in OPZ-18 suggests this unit behaved similar to the underlying unit, MP-2d in the area of that borehole.

MP-2d under the central crater area shows the most significant densification. However, densification cannot account entirely for the roughly 50 percent thinning of the unit over a wide area.

Densification within MP-3 accounts for most of the thinning observed within that unit except beneath the central crater. Beneath the central crater, collapse of vugs could compensate for considerable volume loss (thinning) with little observed density increase.

MP-4a shows essentially no densification under the central crater area and at least 40 percent thinning over a wide area. MP-4a and MP-4b are differentiated by organic concentration. It is probable that both units flowed and mixed obscuring their relationship in a manner so that the organics identifying MP-4b occur higher than predicted, and thinning in MP-4a is exaggerated (especially in OPZ-18). Nevertheless, the whole unit, MP-4, experienced 20 percent thinning over a wide area that is not accounted for by densification.

If the geologic pre-shot models are correct, than the stratigraphic units that show thinning is excess of that explained by densification must have been partially removed by flow. Two units under the central crater area that indicate significant flow are MP-2d and MP-4a. Material from these units previously has been shown to have been piped (vertical flow) to the surface. Material from MP-4 is involved in the majority of deep piping, but the estimated volume of that material preserved in the crater fill only accounts for a small amount of material that flowed.

It appears most of the volume lost is accounted by lateral flow. Two lines of evidence support this: (1) The density increase detected in MP-4 at the base of OTG-23 suggests lateral densification presumably from lateral

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flow, and (2) The uplifted MP-3 over much of the lagoon (refer to fig. 7-21) appears to be caused by thickening of MP-4. This bulging or bulking of MP-4 clearly is visible in all the seismic-reflection survey profiles that run through the lagoon opposite the crater.

Piping (vertical flow) is clearly a post-dynamic phenomenon. The units that appear to have experienced lateral flow, also were involved with latestage piping. These units in OAK appear to be MP-2c over a limited extent (because it was excavated in the central portion of the crater) and MP-2d, MP-4a, and MP-4b (over a wider extent). Deep piping appears to have been vented initially through the central part of the crater and subsequently through concentric fracture zones developed farther laterally (due to subsidence) represented by the piped mounds or volcanoes preserved on the terraces today. Shallow piping appears to be a more widespread phenomena over the crater area, though it too was, at least partially, vented through the central part of the crater as evidenced by the presence of shallow-piped material in all central crater zones.

# KOA Crater.

By comparison with OAK crater, KOA experienced much more shallow lateral and vertical flow within its units. Thinning and bulking (thickening) is observed in MP-2b, MP-2c, and MP-2d. The thinning appears in increasing area in each lower unit and, therefore, the thickening occurs farther from groundzero with depth. MP-3 appears more thinned than in OAK and may have experienced flow. MP-4 appears to have thinned and flowed. However, the thinning in MP-4 is complicated in its area of distribution; represented by central rebound and perhaps channeled vertical flow through the central uplift. Shallow piping appears to have been pervasive and is indicated by the common EE material in much of the mixed zone. Venting of some of the shallowpiped material in the areas of KBZ-4 and KCT-5 is suggested by the deeper mixing of this shallow-piped material in the crater fill of these boreholes.

#### Relative Timing of Depositional Events

The idealized succession of depositional events is shown in Figure 7-33.

For purpose of discussion, the following stages of crater development, referred to as craters, are defined:

- (1). <u>Maximum transient crater</u>. That crater formed when the outgoing velocity vector is zero, prior to rebound. The formation of the Beta 3 rubble beneath the excavational crater is thought to represent the maximum transient-crater growth.
- (2). <u>Terminal transient crater</u>. That crater formed at the end of the transient-crater phase, following rebound.
- (3). <u>Collapse crater</u>. That crater formed at the end of the formation of the rubble from the collapse of the sidewall/flap (early stage collapse, Beta lb).

- (4). Initial slump crater. That crater formed at the end of the formation of late-stage collapse rubble (Beta la).
- (5). <u>Apparent crater</u>. That crater observed today, determined by extrapolation of post-shot measurements to the land or water surface at shot time (B. L. Ristvet, personal communication). This crater also has been referred to as the subsidence crater.

For OAK crater, in chronologic order from oldest to youngest, the sequence of depositional events is:

- (1). Formation of transition sand (Beta 2) at base of transient crater. dynamic lateral flow of subsurface units, air-blast deformation.
- (2). Collapse of excavational-crater wall/rim destroying lateral extent of transition sand and forming collapse rubble (Beta 1b); formation of collapse crater; initiation of liquefied (post-dynamic) flow, especially in MP-2c, MP-2d, MP-4a, and MP-4b; and initiation of subsidence.
- (3). Penecontemporaneous formation of undifferentiated rubble zone external of collapse crater by partial flap collapse in addition to prior airblast deformation.
- (4). Penecontemporaneous formation of debris blanket by probable partial failure and movement of the excavational-crater wall/rim lagoonward.
- (5). Infilling of at least part of remaining crater bowl (collapse crater) by wash- and/or fall-back initiating deposition of "hiatus" sand (Beta ls); and initiation of winnowing (removal in water suspension) of fine-grained sediments.
- (6). Continuation of deposition of "hiatus" sand over outer crater and contemporaneous initiation of shallow piping in the central crater; continued liquefied lateral flow and subsidence; continued winnowing.
- (7). A sequence of crater-margin collapses to form graded rubble (Beta 1a). One collapse (slump) resulted in a flow large enough to cross the crater, breach the debris blanket, and flow into the lagoon. Subsidence, piping, liquefied lateral flow, and winnowing continued. Shallow-piped material reached the surface throughout the crater before or during initiation of Beta 1a deposition.
- (8). Margin slumping and graded-sand (turbidite) deposition (Alpha 2); subsidence, piping, liquefied lateral flow, and winnowing continued. Deep-piped material reached the surface at the beginning of Alpha 2 deposition.
- (9). Late-time partial infilling of central part of the crater with mud (Alpha 1); subsidence, piping, and lateral flow continuing but progressively less. Mud deposition has continued to present. Local slumping, sand deposition, and winnowing has continued along the reef margin mainly as a consequence of natural geologic processes.

Because the OAK crater developed on a slope, the bathymetric center of the crater moved downslope at the end of early-stage collapse (generally shown as an apparent, progressively lagoonward migration of the low point of successive crater-fill units in fig. 7-25B).

For KOA crater, the events (from oldest to youngest) are:

- (1). Same as in OAK.
- (2). Same as in OAK; liquefied lateral flow especially in MP-2b, MP-2c, MP-2d, MP-4a, and MP-4b.
- (3). Penecontemporaneous formation of debris mounds by partial MIKE-ward collapse and movement of crater wall/rim.
- (4). Same as in OAK, resulting in the much-thickened section at KET-7.
- (5). Same as in OAK.
- (6). Possible deposition of a thin "hiatus" sand over outer crater (this was mostly destroyed by subsequent collapse); initiation of shallow piping; and continued liquefied lateral flow and subsidence.
- (7). A sequence of crater-margin collapse to form graded rubble; subsidence, shallow piping, and liquefied lateral flow continued.
- (8). Margin slumping and graded-sand (turbidite-flow) deposition (Alpha 2) and piping in the central part of the crater; continued but reduced subsidence and lateral flow. Deep-piped material reached the surface at the initiation of Alpha 2 deposition.
- (9). Late-time infilling of central part of the crater with mud (Alpha 1); localized slumping and sand deposition around most of crater (except near MIKE). Subsidence has lessened markedly. Deep piping has discontinued before deposition of mud (Alpha 1).

# **VOLUME PROBLEMS**

Beyer (Ch. 2 of this report) and Trulio (Ch. 6) demonstrate that densification can only account for 8 to 15 percent of the subsidence measured in the crater wings of OAK. Yet OAK appears to be substantially a subsidence crater. Peterson and Henny (Ch. 5) show substantial late-time subsidence (post two months post-shot) by comparing the 1958 H&N post-shot map with the 1984 USGS bathymetric map. Peterson and Henny estimate an apparent (conservative) volume increase of the crater of 231 million cubic ft or 25 percent of the apparent crater volume. Piping clearly demonstrates the existence of a long-term unstable liquefied mass at depth beneath the crater.

Piping is <u>one</u> avenue by which this liquefied mass could achieve stability. It is the expression of liquefied flow vertically. Dikes, lagoon uplift, and densification of MP-4 laterally all demonstrate lateral flow.

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# Evidence for Piping and Lateral Flow

Piping is not an hypothesis but a process the results of which are observed in both OAK and KOA craters.

<u>Surface. --</u> Sand volcanoes are present on the terraces of OAK crater (fig. 7-14). Their presence on the terraces indicates that deep piping persisted after the majority of Alpha deposition (slumps and sand turbidity flows) and that flow (at least vertical) is a very long-term process.

Crater-Fill. -- The volume of piped material is estimated for the craterfill of the central bowl of OAK. Deep-piped material (biostratigraphic zones II-MM) is 4.83 million cubic ft. Shallow-piped material (biostratigraphic zones FF-GG) is 45.62 million cubic ft. Deep-piped material is only surficial outside the central bowl (i.e., on the terraces). Shallow-piped material is throughout the crater-fill above the hiatus sand (Beta 1s), so an appreciable additional volume exists but has not been calculated (further paleontologic study is needed for these estimates). Shallow piping reached the crater surface after deposition of the hiatus sand. Deep piping reached the crater surface after deposition of the graded rubble (Beta 1b). This is true for both OAK and KOA craters and suggests that shallow piping reached the surface on the order of minutes, deep piping on the order of hours, and both persisted for days. In OAK, deep- and shallow-piped material is present in the latestage mud (Alpha 1) deposits suggesting very long-term deposition (many months). In KOA, only shallow-piped material is present in the late-stage mud (Alpha 1) deposits. Deep-piped material is present in the graded sands (Alpha 2) suggesting an ending of deep-piping at the initiation of Alpha 1 deposition in KOA (approximately 1-2 months). If the liquefied material mass exhibits vertical flow for days, why is lateral flow constrained to the dynamic phase of crater development? Material from MP-4 (deep-piped) would have an effective seal by MP-3 in most places except in the proximity of large fractures and vents. The material was depressed under the central crater and the sealed avenue of flow would be up along the lower surface of MP-3.

Subsurface. -- Unit MP-4 shows densification away from the central crater. Borehole gravimetry shows a substantial increase in density in MP-4 in OQT-23 for that part that was measured (refer to tbl. 7-9). Gamma-gamma density shows a similar substantial increase in density throughout MP-4 in OKT-13. The H&N post-shot bathymetry (Peterson and Henny, Ch. 5 of this report) shows a rise in the lagoon sea-floor depth that exceeds the thickness of the debris measured in several boreholes. This indicates uplift of the surface. As previously mentioned, MP-3 shows uplift throughout the lagoon (refer to fig. 7-22). Both these lines of evidence suggest significant lateral flow of MP-4, densifying the material laterally and uplifting the overlying sediment and rock over much of the lagoon. It appears that calculations down to the line of zero net displacement (contour D, Trulio, Ch. 6 of this report) do not adequately encompass all significant subsurface crater phenomena because much appears to happen below this line.

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Loss of Material from the Crater. -- In OAK, large ejecta or debris is found entirely within the apparent crater. Halley and others (1986b) demonstrated several clasts (unspecified size, but generally cobble-sized or larger material collected by submersible) to be debris. Debris rays were observed along the reef tract, generally all within the apparent crater. The debris blanket is within the apparent crater. Where did all the material go to form the apparent crater?

Debris is indicated outside the apparent crater in three places:

- In the breach and turbidite flow deposit (fig. 7-14) deposition of an unknown volume of debris is inferred from the enhanced sea-floor image.
- (2). At OAR-2A, the upper 25 ft of sediment (mostly sand) contains highly shocked material, probably ejecta swept from the reef tract and deposited at this site. This thickness could represent a substantial amount of material deposited along the toe of the slope from reef to lagoon on both the northeast and southwest sides of the crater.
- (3). In mud in the lagoon, clay-sized material is common in samples from throughout the northeastern part of the lagoon (Wardlaw and Henry, 1986b). It is probably mostly blast-derived and represents a substantial volume of fine-grained ejecta and material from the craters. Post-shot photographs show mud-ladened plumes into the ocean and lagoon far beyond the apparent crater, suggesting this loss was not trivial.

The uplifted area in the lagoon near OAK (fig. 7-22) suggests that a considerable volume of material from MP-4 moved outside the apparent crater in the subsurface. If a 5-7 percent density increase accompanied this thickening it would account for an appreciable amount of the apparent crater.

#### SUMMARY

OAK and KOA craters are similar. They exhibit the same geologic crater zonation. The zone of sonic degradation that defines the geologic crater is very similar for both craters. OAK and KOA differ in type of device, in coupling, and in depth and radius of the various stages of crater development that are not within the scope of this paper. The KOA area was preconditioned by MIKE and possibly other devices. It contains a better, thicker cemented interval (MP-3) at shallower depths than OAK. These two factors contributed strongly to the major differences between KOA and OAK. KOA is a crater that developed early and had far less late-stage modification, as indicated by its lack of late-stage piping and diminished late-stage subsidence. OAK, on the other hand, is a crater most of which developed later and had significant late-stage subsidence and piping. It appears that as much as 66 percent of the apparent crater volume of OAK may be due to subsidence. In contrast, only about 20 percent of KOA may be due to subsidence.

Piping requires a liquefied material mass at depth. Piping lasted for months at both craters. Subsidence lasted for months at both craters. The prolonged existence of a liquefied material mass at depth is related causally to prolonged subsidence.

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## OAK CRATER H&N (1958), POSTSHOT CON







## **DSTSHOT CONTOUR MAP**



## OAK CRATER H&N POSTSHOT - H&N PRESHOT ISOPACH





# OAK CRATER ESHOT ISOPACH MAP, NEGATIVE &-RELIEF

#### OPEN-FILE REPORT 87-665 PLATE 5-4





PLATE 5-4 OAK CRATER, ALICE REEF (STA.25) EARLYTIME NEGATIVE △-RELIEF ISOPACH MAP

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#### DEPARTMENT OF THE INTERIOR U.S. GEOLOGICAL SURVEY

## OAK CRATER H&N POSTSHOT - H&N PRESHOT ISOPAC



















**OPEN-FILE REPORT 87-365** 

PLATE 5-8

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OPEN-FILE REPORT 87-665 PLATE 5-9

PLATE 5-9 OAK CRATER, ALICE REEF (STA.25) COMPARATIVE POSTSHOT ISOPACH MAP (POSITIVE -RELIEF)

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