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GEOTECHNICAL ASSESSMENT OF UNITED STATES
AND FOREIGN TEST SITES AND MATERIAL PROPERTIES
OF GEOLOGIC MEDIA

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ABSTRACT

This report presents a geotechnical assessment of geologic materials. This geotechnical assessment includes a compilation of both physical and mechanical property data for a wide variety of materials of interest to developing *in situ* constitutive relations for a variety of geologic environments. The data for shale, sandstone, granite and alluvium is presented. The *in situ* stress state was also defined for a variety of geologic and structural environments over a range of depths. An extensive experimental program was conducted on material from the GASSBUGGY site. These include data on the Picture Cliff sandstone and the Lewis shale. Triaxial tests including hydrostatic and uniaxial strain tests were conducted. Problems studied include the intact versus residual failure envelopes, the dilatent behavior of the shale and the sandstone and the dependency of modulus on strain-rate. These latter dynamic tests require the development of new experimental equipment.

Salt was also studied in detail. Cures from both the GNOME event and SALMON event were tested under triaxial conditions; both compression and extension failure envelopes were determined. The results indicate that the salt is ductile and shows no increase in strength beyond 0.7 kilobars stress difference for confining pressures up to 4 kilobars. The stress-strain response of a visco-elastic model was developed and used as the basis for analyzing the data from the dynamic tests conducted on the Lewis shale and Picture Cliff sandstone. The model is presented along with the dynamic data.

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1.0 INTRODUCTION

The objectives of the current program conducted for the Advanced Research Projects Agency and the Defense Nuclear Agency were to evaluate, analyze and develop material properties of geologic media for use in the calculation of reduced displacement potentials and to conduct a geotechnical analysis of specified tests sites in both the United States and foreign countries.

The geotechnical studies have included detailed analysis of sites in the United State and foreign countries. The foreign sites have been analyzed in cooperation with the U.S. Geological Survey, Reston, VA, especially Jack Racklin and William Dempsey. These test sites have been analyzed with respect to lithology and geologic structure. Material properties for several foreign sites were estimated based on data from analogous United States test sites. On the basis of all these data, *geologic models for the sites were formulated*. These data were reported in technical report entitled "Estimated Material Properties of U.S.S.R. Test Sites" (Classified).

Analysis of the current state-of-the-art with respect to testing rock in the Soviet Union was also investigated. Analysis of the U.S.S.R. paper "Statistical Principals of the Strength and Deformation of the Rocks in Complex States of Stress" by A.F. Stavrogin was reviewed to get a current feeling of a state-of-the-art of rock mechanics in the Soviet Union. In addition, a number of American scientists, who have been recent visitors to the Soviet Union, were contacted. Some Soviet scientists from the Russian Academy of Sciences visited Terra Tek on two separate occasions to discuss rock mechanics testing. Their techniques were discussed at that time. The appropriate ASTM standard test methods

were analyzed and discussed for possible use in the Threshold Test Ban Treaty (TTBT) and Peaceful Nuclear Explosion Treaty (PNET) negotiations with the Russians as standard procedures for determining both the tensile and compressive strengths of rock and for obtaining physical properties such as density, porosity, etc.

Dr. Pratt has also been participating in the working group of the ARPA seismic review panel and has discussed the verification and use of data to be exchanged under the Threshold Test Ban Treaty and Peaceful Nuclear Explosion Treaty. A number of meetings were held to delineate the packages to be exchanged between the United States and U.S.S.R. and to analyze the various potential test sites. The first section of the report includes a summary of physical and mechanical property data on geologic material of interest. Testing techniques, outlined in the first section of this report, are currently being used to obtain physical properties that will be exchanged during the Test Ban Treaty. These include density, porosity, and water content. Preliminary cross plots of physical and mechanical property data, density, porosity, water saturation and compressional and shear wave velocities are included. In addition, a section on the role of *in situ* stress and its potential effect on the seismic signal from a nuclear event is discussed. The relation between *in situ* stress magnitudes and depth, age and tectonic environment is evaluated.

Terra Tek has been conducting a program of material properties tests from the GASBUGGY site to supply input data and material models for use in the calculation of reduced displacement potentials by Dr. Robert Bjork and Dr. Robert Allen (PACTECH). Material property tests that have been

conducted include hydrostatic, uniaxial strain and standard triaxial tests on both the Lewis shale and Picture Cliffs sandstone. The tests were conducted at moisture contents simulating the *in situ* environment. In addition, several load path tests were conducted following the load paths indicated by the calculations from PACTECH.

Several problem areas were discussed with PACTECH and it was decided to conduct tests to address ourselves to these areas which significantly affect the results of the Reduced Displacement Potential Calculations. The problem areas include (1) initial versus residual failure envelopes, (2) dilatancy, and (3) strain rate effects on modulus. These tests will enable us to delineate the respective influences of the static and dynamic moduli during loading and unloading in the seismic frequency range. The test procedure loaded the specimen in a one millisecond rise time to a few bars by dropping a weight on to the specimen and analyzing the stress relaxation with time using a laser interferometer. The problem areas delineated above have been studied and data and analysis is presented.

2.0 GEOTECHNICAL ASSESSMENT OF GEOLOGIC MATERIALS

This summary provides some physical and mechanical properties of selected earth materials. It is anticipated that these data taken from published results of laboratory and selected field tests will provide a "data-base" for those developing constitutive equations of geologic materials, calculating the response of geologic materials to some loading, conducting experiments in earth media, or considering soil or rock/structure interaction. The geologic materials selected include:

granite
sandstone
shale
alluvium.

These materials were selected because of the existing data base and because existing foreign sites are situated in these materials. In addition, tests on shale, sandstone and salt have been conducted under this contract and are presented in a later section.

The material property data include a description of the "physical" nature of each material and the "mechanical" response to a variety of load paths. The physical properties include some indication of the mineral composition, the density, the porosity and the grain size. The mechanical properties include the general quasistatic-room temperature stress-strain response up to stresses of the order of 6-10 kilobars, with limited hydrostatic compression data. Some high strain-rate tests and ultrasonic test data are also included.

Rock and Soils Included:

"Granite"*

Westerly granite (Rhode Island)

Climax Stock granodiorite (Nevada Test Site)

Sandstone

Kayenta formation (Mixed Company site, Colorado)

Nugget formation (Utah)

Shale

Pierre formation (Middle Gust site, Colorado)

Pierre formation (South Dakota)

Lance formation (El Paso Wagon Wheel site, Wyoming)

Nevada Test Site formation (Yacht site, Nevada Test Site, Nevada)

Alluvium

Yucca flat alluvium (Yucca flat, Nevada Test Site, Nevada)

Westerly Granite

The Westerly Granite is a fine-grained igneous rock, grayish pink in color and composed of quartz, potassium feldspar and plagioclase feldspar, with minor amounts of biotite and hornblende. Grain size averages about .75 mm. Porosity is less than one percent and permeability is very low.

*Granite rock here includes a broader based group of rocks than the usual granite as defined by the geologist, based on mineralogy. The group would include granite, monzonite, granodiorite, and diorite whose mineralogy differ, but whose mechanical properties will probably be similar.

Climax Stock Granodiorite

This rock is medium-grained, grayish white, with grain size ranging between .5 to 5 mm. It is composed of quartz, potassium and plagioclase feldspar and micas. The rock has a typical interlocking igneous texture resulting from crystallization from a molten state. Both porosity and permeability of the intact rock is very low.

Nugget Formation

The Nugget Formation is a fine-grained, buff colored sandstone composed primarily of quartz, minor feldspar and hematite. The cement is secondary silica. Grain size ranges between .1 to 1.3 mm. Porosity is low for a sandstone, approximately 4 percent.

Kayenta Formation

Kayenta Formation is a fine to medium-grained porous sandstone, buff colored to almost white, and is composed of subangular to rounded grains of quartz, feldspar, rock fragments and calcite. The cement is predominantly calcite. Minor opaques and micas are also present. Porosity ranges between 15 to 25 percent. Bedding thickness ranges from a few inches to several feet.

Pierre Formation - Middle Gust Test Site, Colorado

The Pierre Formation is a soft fine-grained, gray to black clay shale with stringers of gypsum and sand size material. The clay content is very high. Pyrite is sometimes found in the clay shale unit. Bedding thickness is highly variable.

Lance Formation

The Lance Formation is a shale to silty-shale, medium to gray to black in color with light gray stringers of coarser silt and sand sized material. The coarser grained stringers have the texture and composition of a fine-grained sandstone and indicate that the variation in lithologic character sometimes occurs only over a very few millimeters distance. These stringers permeate the shale at orientations running from sub-parallel to perpendicular with respect to the primary bedding which is oriented perpendicular to the core axis.

Yacht Site Shale

The Yacht Site shale is a Paleozoic fine-grained black shale having contorted bedding planes, which vary in orientation over small distances. Silt size grains of quartz were very common in several layers, also some fine-grained carbonate is present.

Pierre Formation - Shale

The shale is part of the Cretaceous Pierre Formation and is a grayish-black, fine-grained shale with a high clay content. Bedding is usually thin and the rock is sometimes very fissile.

Yucca-Flat Alluvium

The alluvium is a weakly cohesive, poorly sorted clastic material composed largely of quartz, feldspar, clay minerals and rock fragments. Some carbonate may be present. Grain size ranges from silt size particles to cobbles a few centimeters in diameter.

2.1 Physical Properties

The physical properties for the materials of interest are present in Table 1. Table 2 summarizes the testing techniques and accuracies for the physical property methods. Accuracy was obtained from calibration tests. Error was calculated using average Nevada Test Site tuff values:

($\rho_g = 2.40 \text{ gm/cm}^3$, $\rho_w = 1.8 \text{ gm/cm}^3$, $\rho_d = 1.45 \text{ gm/cm}^3$, and $w = 19\%$).

Moisture Content. The moisture content is determined by weighing a sample before and after oven drying with a resulting accuracy of 0.13% H₂O due to loss of water during crushing.

Bulk Density. Dry or "as-received" bulk densities may be found from the volumes determined using (1) calipers, (2) the Ruska Mercury Porometer or (3) the water immersion method. If a precision machined sample is not available for caliper measurement, then for samples 30 cm³ or larger, the Ruska Mercury Porometer should be used, whereas the water immersion method is better for volumes less than 30 cm³.

Grain Density. The National Bureau of Standards powdered silicon (SRM 640) and the crystalline quartz tests indicate that the Beckman Gas Pycnometer technique is more accurate and precise than the water immersion method (0.001 gm/cm³ as compared to 0.010 gm/cm³ respectively). Materials containing zeolites may lead to volume drift in the gas pycnometer unless the sample is allowed to equilibrate with atmospheric moisture prior to testing.

Total Porosity. The total porosity is calculated from the grain and dry bulk density values and consequently depends upon the techniques used

TABLE 1: PHYSICAL PROPERTIES

Name	Rock Type	Composition (Moles %)	Grain Size (mm)	Density		Porosity (%)	Moisture Content (wt. %)	Source
				Grain (gm/cc)	Grain (gm/cc)			
Hesterly	Granite	28 Q ⁺ , 35 K, 31 P, 5 M	0.75	2.646	2.65	~9	Lab dry	Denver Monument quarry Hesterly R.I.
Climax Stock	Granodiorite	29 Q, 16 K, 45 P, 5 M (19 modes)	0.5-4.0	2.67-2.69	2.67-2.70	<1.0	Dry Saturated	Climax Stock Area 15 Nevada Test Site, Nye Co., Nev.
Hogget fm.	Sandstone	96 Q, 2 K 2 Op	.07-0.3	2.52	2.625	4.0	Dry	Perley's Canyon, Salt Lake Co., Utah
Layunta fm.	Sandstone	70 Q, 10 P 8 Cc, 8 Rf, 2 Op, 2 M	.1-.4	1.91-2.20 1.98-2.01 2.08	2.65	14-20 26.4 21.2	Dry	Mined Company test site Mesa Co., Colorado
Pierre fm.	Shale	Ca, Cc, Q	<.05	2.07	2.71		As received Saturated	Middle Gust test site, Greeley, Colorado
Lance fm.	Shale	33 Q, 34 Ca, 13 Cc, 16 Rf, 2 M	.03-0.2	2.47	2.65	0.0	Dry	El Paso Natural Gas Co. Wagon Wheel Well No. 1, Depth 10215, Sublette Co. Wyoming
Yacht	Shale	10 Q, 80 Ca, 3 Cc, 7 Op	<0.03	2.64	2.75	4.0	Dry	Mole UEL, Nevada Test Site, Nye Co., Nev.
Pierre fm.	Shale	Q, Ca, Op	<0.03	2.11	2.70	22.0	As received (non-sat'd)	U2 Minuteman Site
Tecoma Flat	Alterium	Variable	.1-30	1.57			Dry, 7, 10, 17	Area 3 Mole V03-ct Nevada Test Site, Nye Co., Nev.

Q = Quartz P = Plagioclase feldspar Cc = Calcite Rf = Rock fragments
 K = Potassium feldspar M = Mica Ca = Clay minerals Op = Opacites

Table 2

Accuracy and Error of Physical Property Methods

PHYSICAL PROPERTY	HOW DETERMINED	ACCURACY*	STRUCTURAL INTEGRITY ROCK NECESSARY FOR MEASUREMENT
Moisture Content (w)	Weight determined using electronic balance	0.13% w	poor to good
Wet Bulk Density (ρ_w)	Volume measured using 1) Calipers 2) Ruska Mercury Porometer 3) Water Immersion	1) -.014 to -.028 gm/cc 2) .001 gm/cc \pm .002 gm/cc 3) -.002 gm/cc \pm .004 gm/cc	1) fair to good 2) fair to good 3) poor to good
Dry Bulk Density† (ρ_d)	1) Calc. using w and ρ_w 2) Calipers 3) Ruska Mercury Porometer 4) Water Immersion	1) .004 gm/cc \pm .002 gm/cc 2) -.017 to -.031 gm/cc 3) .004 gm/cc \pm .002 gm/cc 4) .001 gm/cc \pm .004 gm/cc	1) refer to w & ρ_w 2) fair to good 3) fair to good 4) poor to good
Grain Density (ρ_g)	Volume measured using 1) Water Immersion 2) Gas Pycnometer	1) .01 gm/cc \pm .04 gm/cc 2) .001 gm/cc \pm .005 gm/cc	1) poor to good 2) poor to good
Total Porosity** (n_t)	Calculated from ρ_g and ρ_d	$-0.14\%n_t \pm .21\%n_t$	fair to good
Effective Porosity (n_e)	1) Ruska Mercury Porometer 2) Water Immersion 3) Gas Pycnometer	0.53% n_e	1) fair to good 2) poor to good 3) poor to good
Permeability (K)	Permeability equipment	Repeatable to within 20%	poor to good
Air Voids** (Av)	Calculated from ρ_g , ρ_d , ρ_w	.16% Av \pm .44% Av	refer to densities
Saturation** (Sr)	Calculated from w, n_t , ρ_w	0.26% Sr \pm .56% Sr	refer to e, n_t , ρ_w
Microscopic Analysis (Grain size, orientation and mineral content)	1) Large grain method using Bausch and Lomb Monocular 2) Medium grain method using Zeiss 3) Medium grain method using universal stage 4) Small grain method using scanning electron microscope	For all microscopic methods accuracy is dependent upon magnifi- cation level and smallest grain size	1) poor to good 2) fair to good 3) fair to good 4) poor to good

* Errors may be calculated using average NTS tuff values ($\rho_g = 2.40 \text{ gm/cm}^3$,
 $\rho_w = 2.80 \text{ gm/cm}^3$, $\rho_d = 1.45 \text{ gm/cm}^3$, and $w = 19\%$)

** Accuracy calculated from Ruska Porometer and Beckman Pycnometer values.

† Dry Bulk density accuracy includes moisture content error.

in those determinations. Using the most accurate techniques, the total porosity accuracy is within -0.35 percent n_t to 0.07 percent n_t .

Effective Porosity. Effective porosity is determined using water immersion, mercury injection (Ruska Mercury Porometer) and gas pressurization (Beckman Gas Pycnometer). The Ruska Mercury Porometer required knowledge of the mean sample pore size for complete effective volume determination. While the gas pycnometer can determine effective pore volume, it cannot easily measure bulk volumes, thus, requiring another method to determine sample bulk volumes. Only the water immersion method is self-contained in that both bulk volume and effective pore volume may be determined using the same piece of equipment. No effective porosity standard exists, but method comparison showed the effective porosity to be repeatable to within 0.53 percent n_e .

Percent Saturation and Percent Air Voids. Enumeration of these properties is subject to accumulated error from density, porosity and percent moisture determinations. The accumulated accuracies for each property are shown in Table 2. The percent saturation and percent air void accuracies are computed using the most accurate techniques for density, porosity and percent water.

Permeability. Permeability may be determined by either a transient or a steady-state method. No generally accepted permeability standards are available, but comparisons and variations in method suggest the values are repeatable to within 20% .

Microscopic Analysis. Microscopic analysis consists of grain size, spatial orientation and mineral content determination. Four methods are available, each of which has its advantages. Mineral content analysis is not possible using the scanning electron microscope (Method IV). However, it provides means of testing low cohesion rock fabrics as does the "large grain method" (Method I). The "medium grain techniques" (Methods II and III) require fair to good cohesive rock fabrics because of thin-sectioning and handling. As stated in Table 2, the determining factors deciding which method to use are the grain size and the structural integrity of the rock.

In order to provide a "feeling" for the relationship between various parameters to physical properties cross plots relating porosity, density, velocity and modulus are given in Figures 1 - 4.

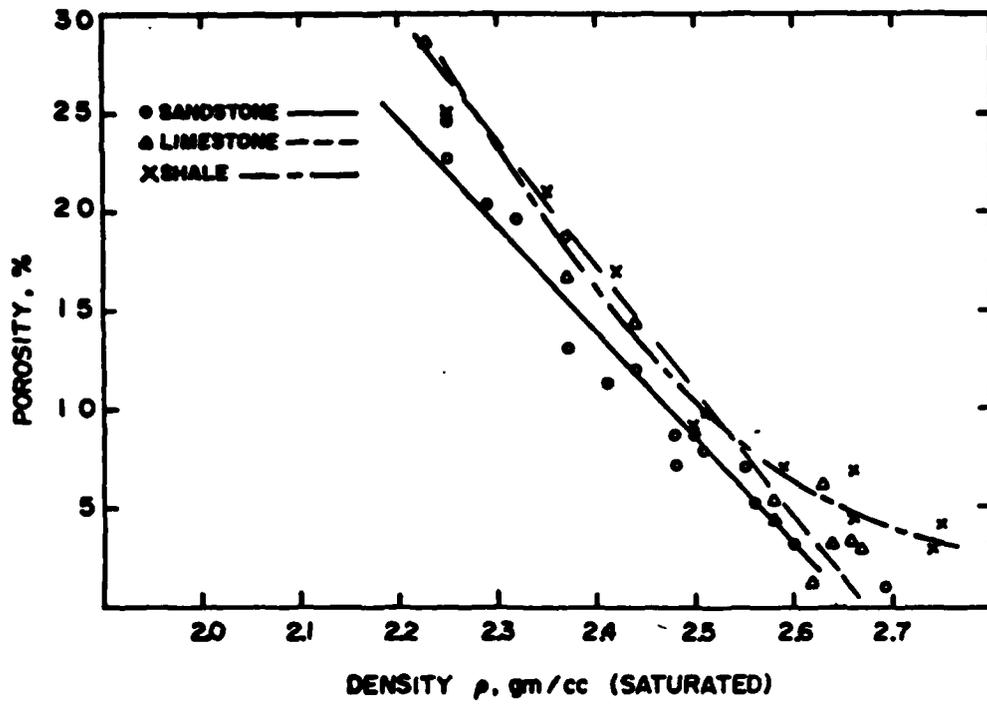


Figure 1. Density vs. porosity (field and lab) - sedimentary rock.

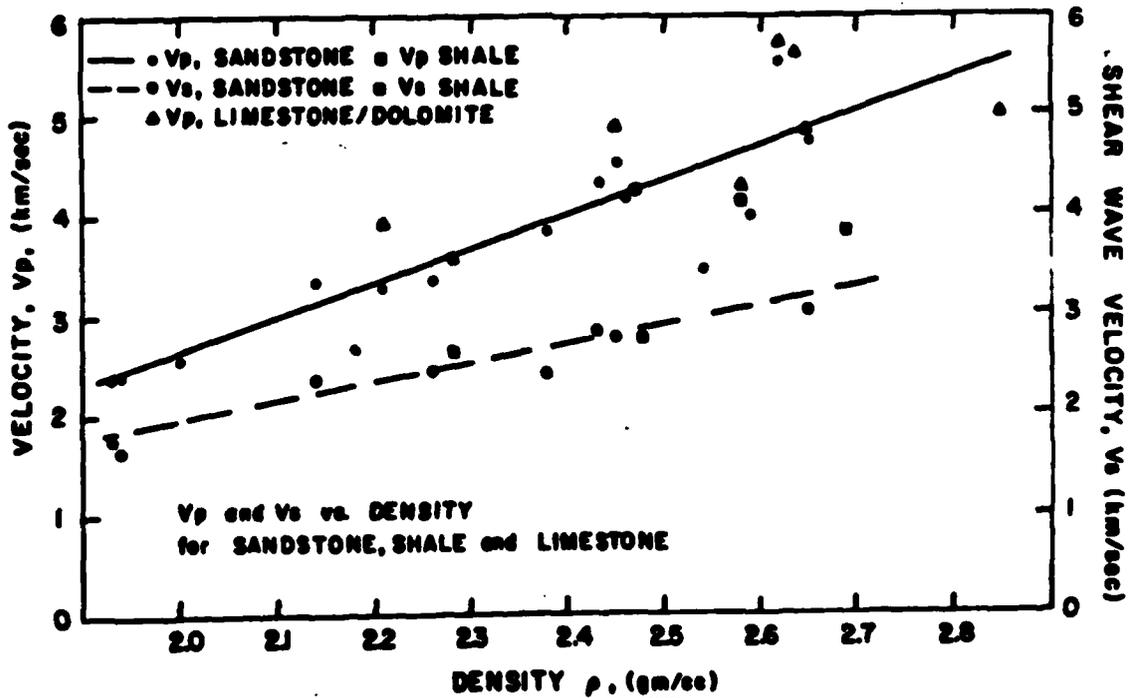


Figure 2. Density vs. V_p and V_s (field and lab) - sedimentary rock.

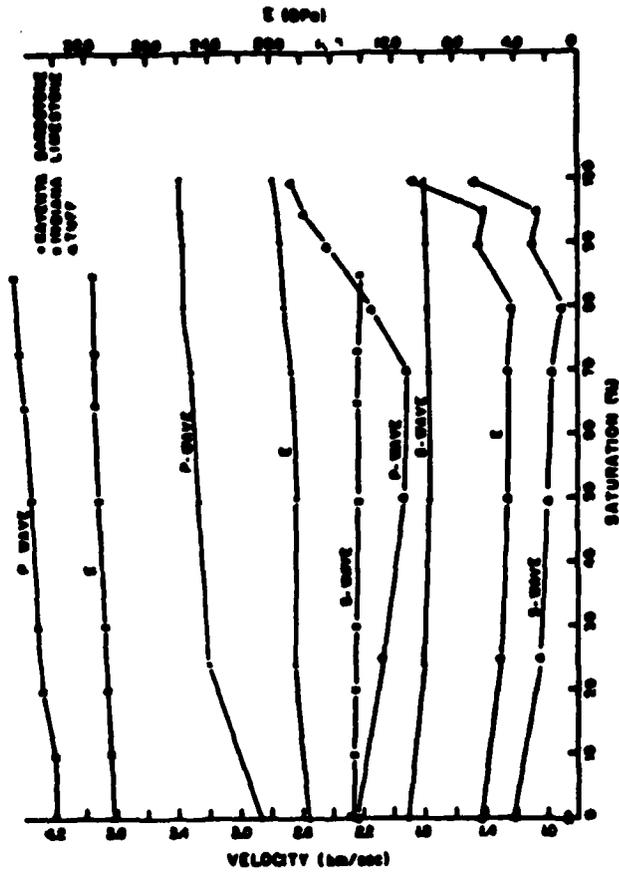


Figure 4. V_p , V_s , and modulus (E) vs. saturation

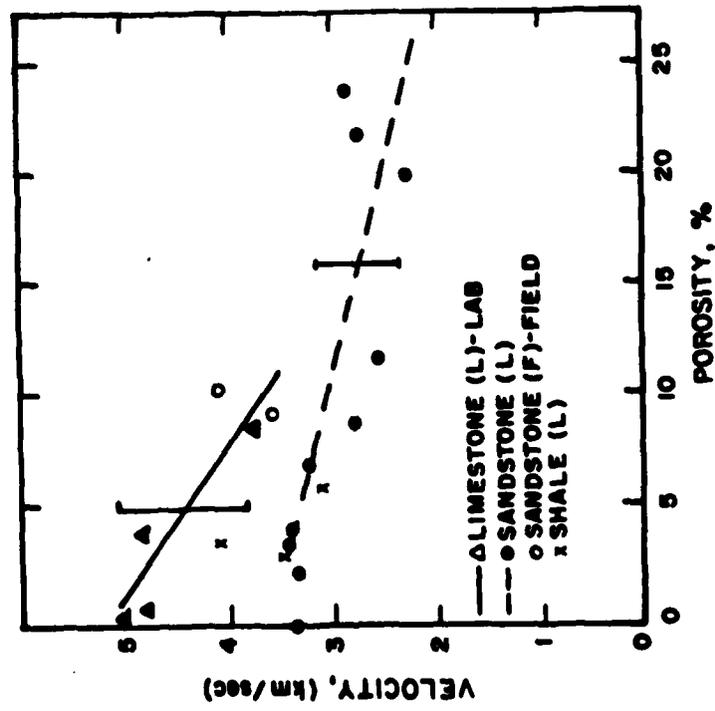


Figure 3. Porosity vs. V_p (lab) - sedimentary rock.

2.2 Mechanical Properties

Laboratory tests have been conducted on geologic materials for some time, however, these tests have to a large extent been restricted to unconfined and triaxial stress tests to high pressure on strong rocks^{2,3} and to constrained tests on soils at low stresses⁴. The development of servo-controlled machines for testing rock in the late 1960's⁵ allowed a variety of load path tests to be conducted ranging from uniaxial strain to proportional loading at stress levels exceeding several kilobars (Table 3). Pore pressure measuring devices are also now capable of applying pressures up to several kilobars. Other static loading apparatus using piston cylinder dies employing a solid pressure transmitting media has been used to static pressure of 40 kilobars⁶. Loads could generally be adequately determined, and strain gages attached to the surface--or to thin metal jackets containing the rock provided strain measurement. Laboratory sample sizes usually ranged from 2.5 to 10.0 cm in length with a characteristic L/D = 2.0. Large loading frames and pressure vessels now allow testing on samples up to 30.0 cm in length to confining pressures of 2.0 kilobars.

Special techniques were required to load and to measure strains for less competent geologic materials such as tuffs, shales, soils, alluvium. Mechanical extensometers were developed in the early 1970's⁷ for adequately measuring large strains. Specimen preparation, and particularly sealing the specimen from confining-fluid intrusion required special development in order to conduct tests to multikilobar stress levels. Sealing can now generally be accomplished for a variety of materials.⁸ Strain rate effects have been studied over a wide range of loading rates from 10^{-5} to 10^{-1} , using static testing equipment⁹ and $10^3/\text{sec}^{-1}$ using shock loading techniques^{10,11}.

To obtain material properties in the 10 to 40 kilobar range, solid media apparatus have been developed to measure the pressure-volume response. At pressures exceeding approximately 40 kilobars, one dimensional shock wave experiments are used to provide most of the information on material properties. These experiments measure either stress or particle velocity in the direction of a propagating wave in the shock front.

Techniques have been developed to measure the frictional properties of jointed material including strength, coefficient of friction and "stiffness" under a variety of loading conditions¹³⁻¹⁶. The various types of shear tests include triaxial, direct and torsional shear tests. Experiments have been conducted with a variety of controlled conditions along the joint or fracture including a variety of surface roughness, pore pressures and joint orientations¹⁷⁻¹⁸.

The data from the laboratory tests are plotted in terms of the principal stress σ_1, σ_2 and σ_3 for the various tests listed in the table below.

TABLE 3

EXPERIMENTAL CONDITIONS		
<u>Test Method</u>	<u>Stress State</u>	<u>Properties Measured</u>
Uniaxial loading	$\sigma_1 > 0; \sigma_2 = \sigma_3 = 0$	Stress-strain behavior; ductility "yield" strength; ultimate stress under static and cyclic loading; effects of temperature and strain rate deformation and fracture mechanisms.
Triaxial compression	$\sigma_1 = \sigma_2 = \sigma_3$	Same as above; pore pressure effects, friction on joints
Triaxial extension	$\sigma_1 = \sigma_2 > \sigma_3$	Same as above
Uniaxial strain	$\epsilon_1 > 0$ $\epsilon_2 = \epsilon_3 = 0$	Same as above
Direct shear Triaxial shear	Multiaxial	Shear strength on prescribed planes including joints; pore pressure effect; friction on joints

* $\sigma_1 > 0$ denotes compression
 $\sigma_1 < 0$ denotes tension

The data for each rock type is presented as a series of plots covering the stress range of interest.

Figures 5 - 9 give the plots available for each rock type covering the range of stress states and strain rates of interest. Average or best curves have been selected for those curves studied by several investigators.

The summary plots include:

- (1) Hydrostats (Figure 5)
- (2) Stress-strain (Figure 6)
- (3) Failure envelopes (Figure 7)
- (4) Uniaxial strain (Figure 8)
- (5) Seismic velocity vs. σ (Figure 9)

These composite plots are presented so that comparisons between various materials can be made.

In addition to mechanical property data, sonic velocity data are presented when available. The apparent elastic moduli can be scaled from the tangents to slopes of the stress-strain curves, and the apparent bulk moduli can be scaled from the mean normal pressures, P_m vs. axial strain curve under hydrostatic loading where:

$$P_m = 1/3 (\sigma_1 + \sigma_2 + \sigma_3).$$

In the deviatoric plane, the square root of the invariants of stress and strain are defined in terms of the principal stresses and strains as:

$$\sqrt{J_2} = (1/\sqrt{6}) \sqrt{(\sigma_1 - \sigma_2)^2 + (\sigma_2 - \sigma_3)^2 + (\sigma_1 - \sigma_3)^2}$$

$$\sqrt{I_2} = (1/\sqrt{6}) \sqrt{(\epsilon_1 - \epsilon_2)^2 + (\epsilon_2 - \epsilon_3)^2 + (\epsilon_1 - \epsilon_3)^2}$$

The relationship between the shear stress and strain, τ and γ , in triaxial tests and the invariants $\sqrt{J_2}$ and $\sqrt{I_2}$ are:

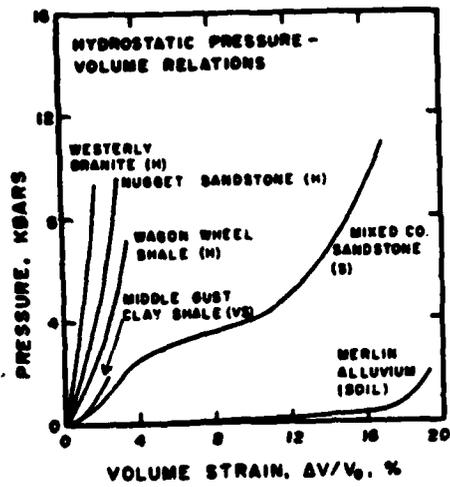


Figure 5. Hydrostatic Pressure - Volume Response of several rocks and alluvium.

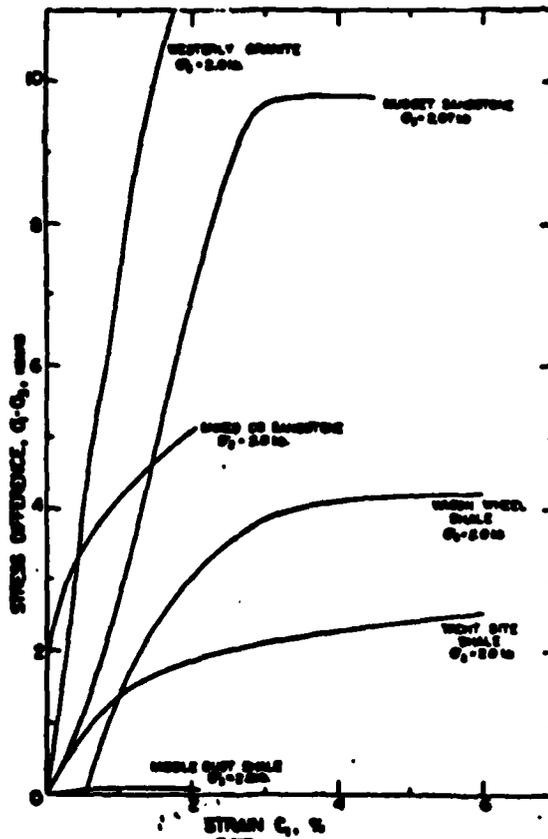


Figure 6. Stress difference as a function of axial strain for shale, sandstone and granite at 2 kilobars confining pressure.

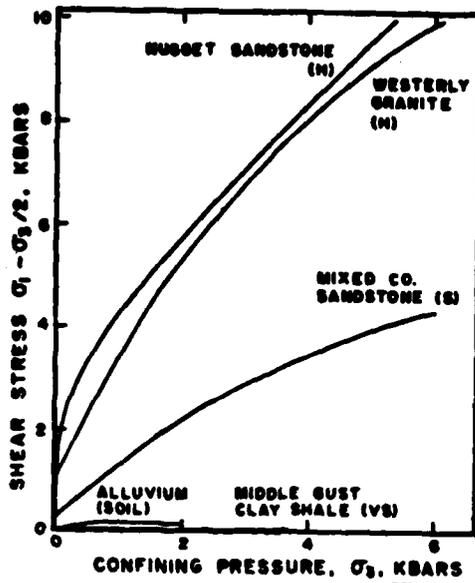


Figure 7. Comparison of the failure envelopes for shale, sandstone, granite and alluvium.

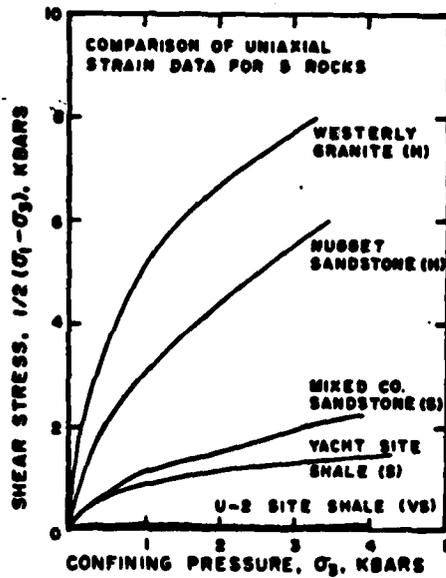


Figure 8. Comparison of the uniaxial strain response for shale, sandstone and granite.

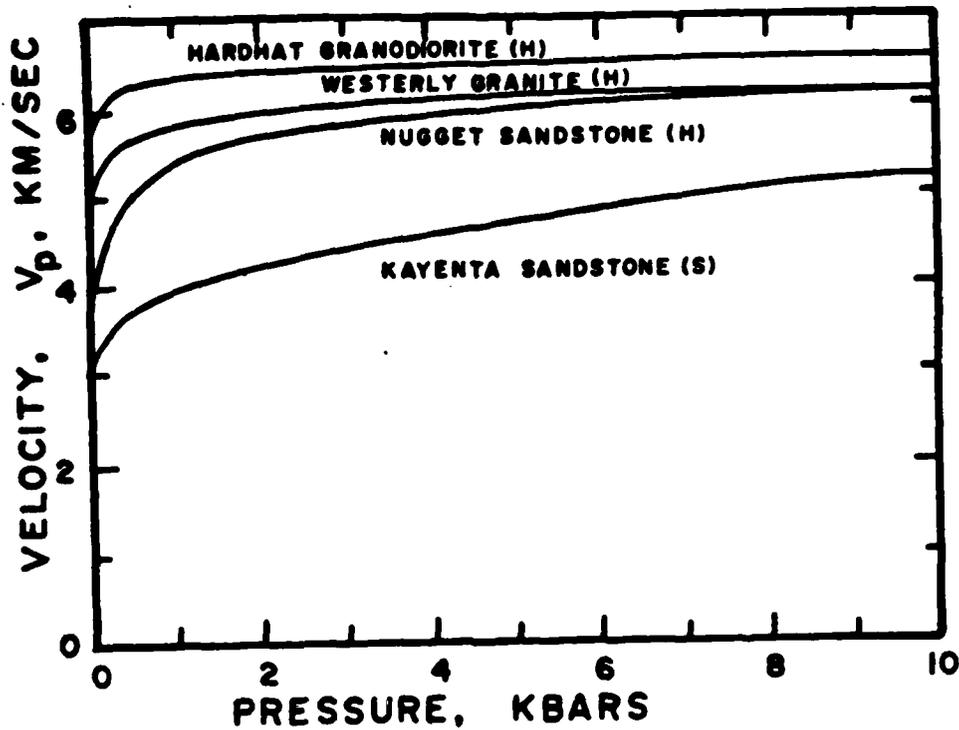


Figure 9. Velocity as a function of pressure for a variety of rocks.

$$\sqrt{J'_2} = (2/\sqrt{3})\tau$$

$$\sqrt{I'_2} = (1/\sqrt{3})\gamma$$

The apparent shear moduli are scaled from the slope of the shear stress-strain curves.

In addition to the coefficient of friction, the frictional properties determined from triaxial or direct shear tests most commonly used in calculations are the shear and normal stiffness. These parameters are scaled from the shear stress-shear displacement and normal stress-normal displacement curves, respectively (Figures 10 and 11)^{19,18}. The roughness of the discontinuity, type gouge, filling material, effect of water and load path all have pronounced effects on frictional strength.

The effect the degree of saturation is also shown for a number of the rocks and soil where data is available (Figure 12)²⁰. In addition, the effect of fracturing and pore pressure on the strength of hard granodiorite is given²¹. Note the magnitude of the decrease in strength between intact and fractured rock and the role of pore pressure to decrease strength (Figure 13). The role of moisture content on failure strength of granite is also given (Figure 14)²². Note the total lack of increased strength with confining pressure with excess water available. Mechanical property data from these rocks as well as Nevada Test Site data on tuff are summarized in Table 4.

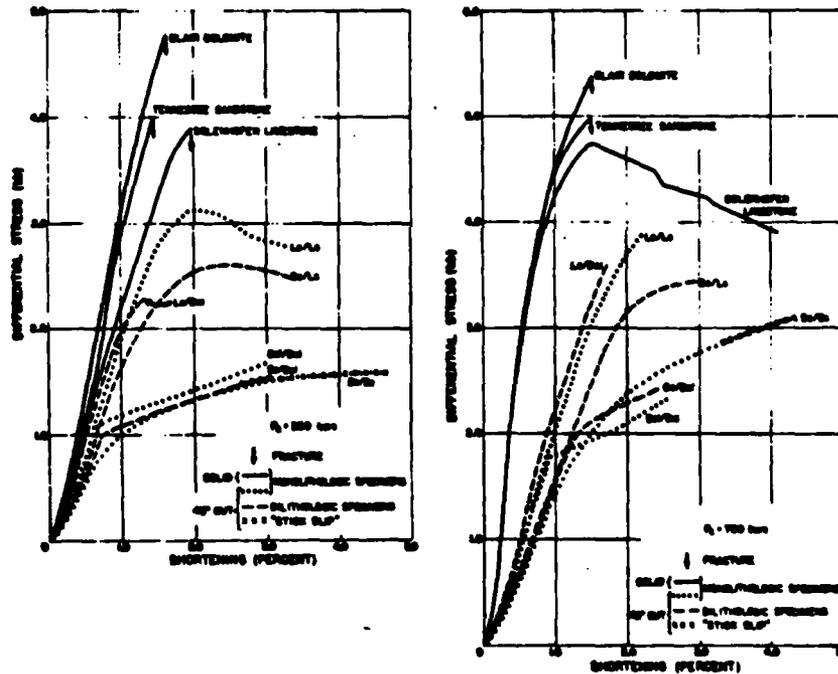


Figure 10. Differential stress vs. axial shortening curves at three different confining pressures for intact specimens (solid curves), monolithologic saw-cut specimens (dotted curves), and dilithologic specimens (dashed curves). Curves are averages of two or more experiments.¹⁹

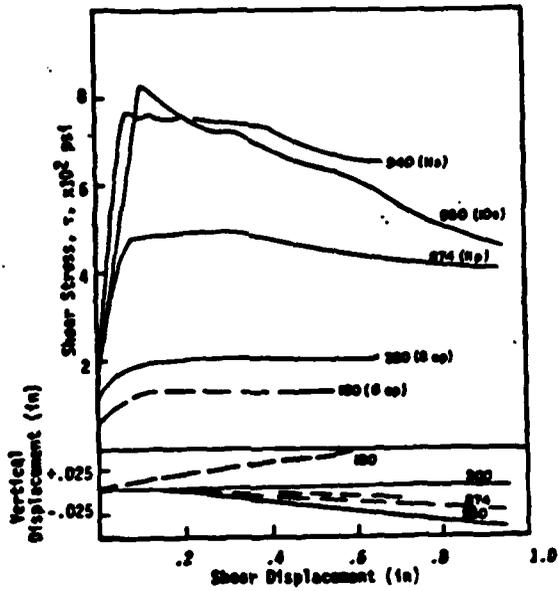


Figure 11. Shear stress vs. displacement curves for shear tests at a variety of normal stresses. From this curve the frictional failure envelope and the shear stiffnesses can be determined.¹⁶

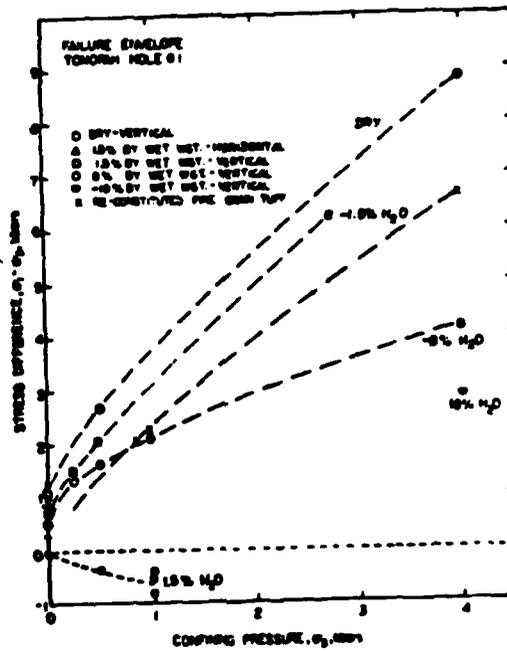


Figure 12. The failure envelope of welded tuff from the Central Nevada Test Site as a function of water content.²⁰

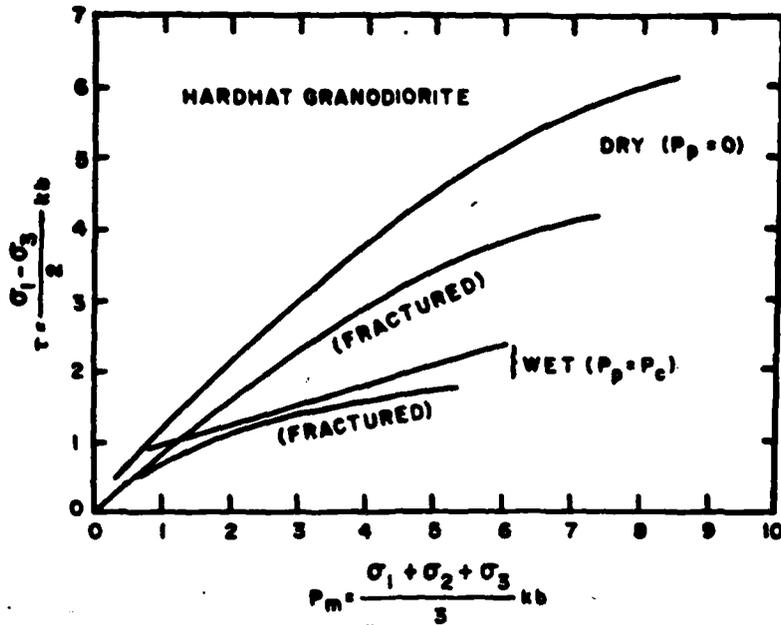


Figure 13. Failure envelopes above intact and fractured Climax Stock granodiorite under dry and saturated conditions 21.

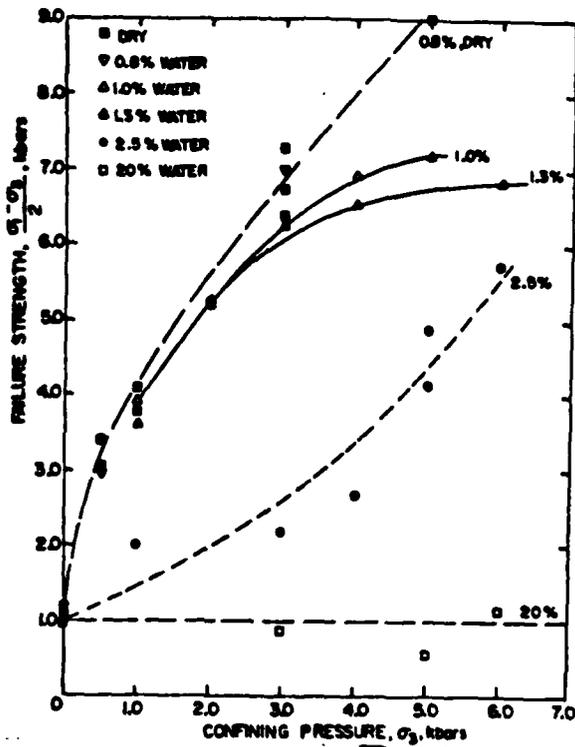


Figure 14. Failure strength of Westerly Granite as a function of confining pressure for dry samples and samples with a variety of water contents²².

TABLE 4

ROCK TYPE (Formation)	UNCONFINED STRENGTH (kb)	TENSILE STRENGTH	STRENGTH DIFF. (kb)	SHEAR MODULUS (kb)
<u>Igneous</u>				
"Volcanic" Tuff Ranier Mesa	0.06 - 0.10	0.01	0.35 - 0.65	4 - 13
Pahute Mesa (more welded)	0.50 - 0.65	0.035		40.0-55.0
"Granitic" NTS Granodiorite Westerly	1.8 2.4	.113		271 227
<u>Sedimentary</u>				
"Sandstone" Nugget Lance (Wagon Wheel) Picture Cliff (Gassbuggy)	2.0 1.5	.117 .054	14.5	140 125
"Shale" Lance (Wagon Wheel)	1.16		8.0	118
"Limestone" Solenhofn	2.8		6.0	262

For the purposes of comparison the published²³ geology (rock type) of Soviet test sites is given in Table 5 as well as the number of events.

TABLE 5

SOVIET PNE SITES	
<u>Rock Type</u>	<u>No. of Projects</u>
Shale/Sandstone	4
Sandstone	9
Limestone & Dolomite	8
Gypsum	1
Salt	3
Granite	7

2.3 In Situ Stress Environment

The role of *in situ* stress on the development of constitutive models for geologic materials is quite evident from nuclear events. *In Situ* tectonic stress can grossly affect the seismic radiation pattern (Figure 15). Seismic radiation is anisotropic and thought to be the result of the release of tectonic stress during the PILE DRIVER and GREELEY events²⁴. The exact contribution of the tectonic stress is not known at this time. In order to better understand the role of *in situ* stress we have plotted the *minimum horizontal stress versus the vertical stress as a function of depth for a variety of materials* (Figure 16). As one can see, in the near-surface environment where NUCLEAR EVENTS are usually conducted, the ratio of the stresses can range from 0.4 to 3, depending on the rock type. Recent data has indicated that more ductile materials such as shales show a minimum to vertical stress on the order of .8 to 1 regardless of depth up to a few kilometers.

Hydraulic fracturing data in crystalline rock and indicates that the stress state varies depending on the tectonic environment (Figure 17). The plot shows environments for normal (N), strike slip (S) and thrust (T) fault areas. The data correlates well with the actual geologic structure in the area. This may enable us to take a first cut of evaluating the stress state for a particular geologic area based on reconnaissance field data. Figure 18 shows the plot of a square root of $\sqrt{J_2}$ (shear stress invariant) as a function of mean pressure. This data indicates that significant deviatoric stresses occur in *in situ* crystalline rock and indeed indicates that granites may be under a tectonic stress environment relatively near the failure envelope. Their data would be exemplified by the *in situ* stress data from the Rocky Mountain Arsenal, where it only took a few

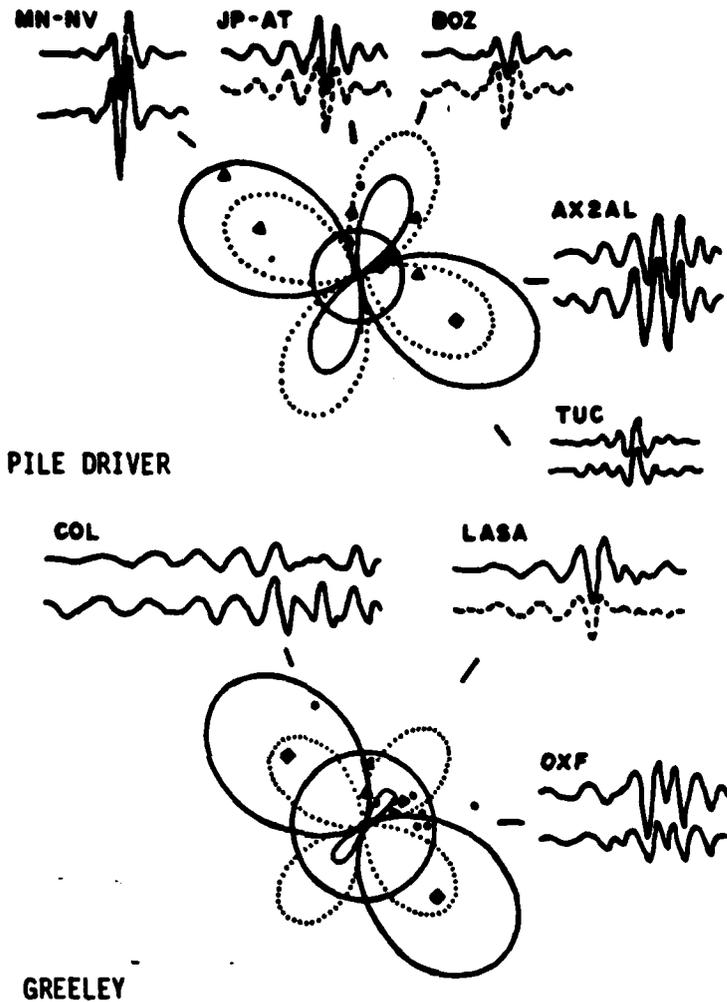


Figure 15. Raleigh wave seismic radiation patterns from events PILE DRIVER and GREELEY showing the anisotropic nature of the radiation. The explosive source (solid circles), strain release component (dashed curves), and composite sources representing theoretical models (solid curves) are shown from various stations (MN-NV, JP-AT, BOZ, AXZAL, TVC, COL, LASA, OXF).²⁴

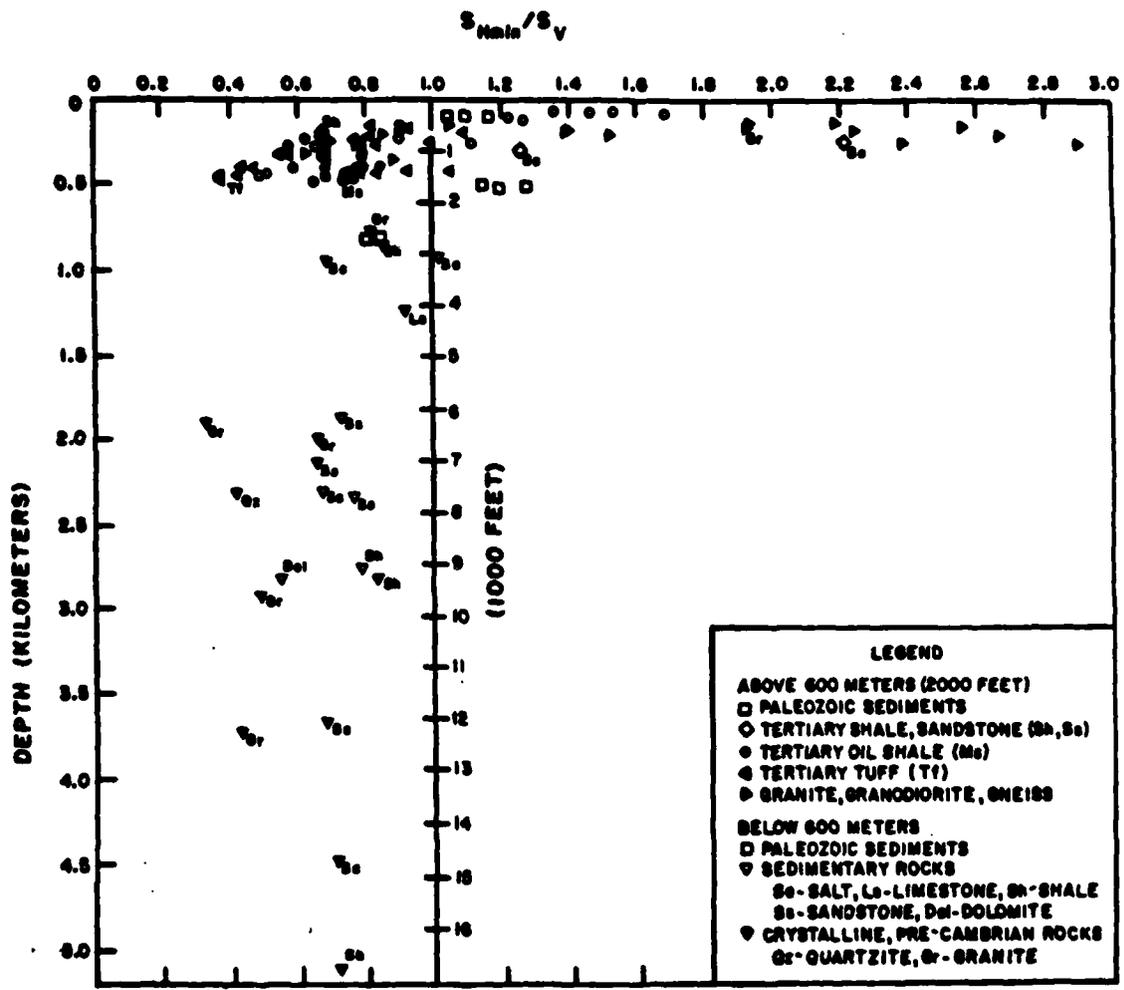


Figure 16. The ratio of the horizontal minimum stress to vertical stress as a function of depth for a wide variety of geologic materials.

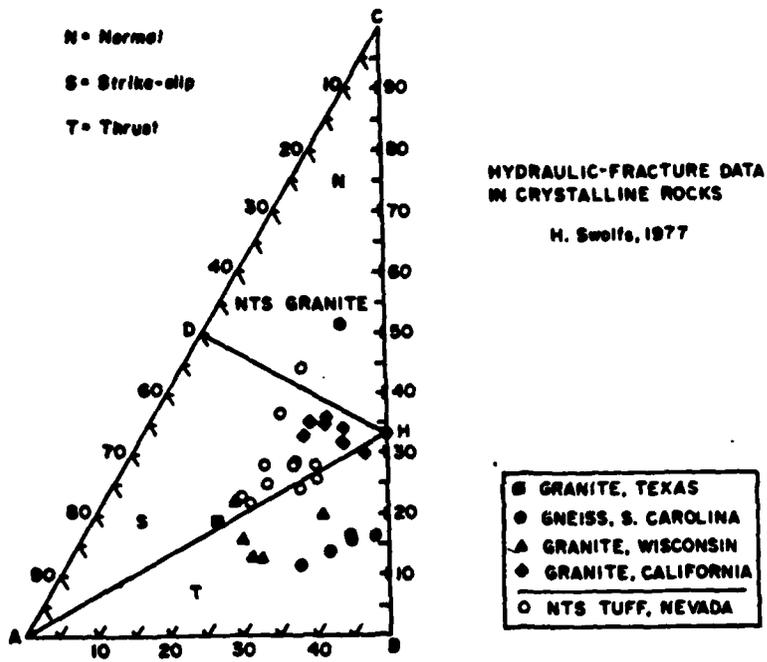


Figure 17. Hydraulic fracturing data in Crystalline rock showing the relationship of the measurements to known tectonic environments. Point H represents hydrostatic loading where $\sigma_1 = \sigma_2 = \sigma_3$. N, S and T represent normal, strike-slip and thrust faulted regions. Corners C and A of the diagram represent σ_1 and σ_3 , respectively.

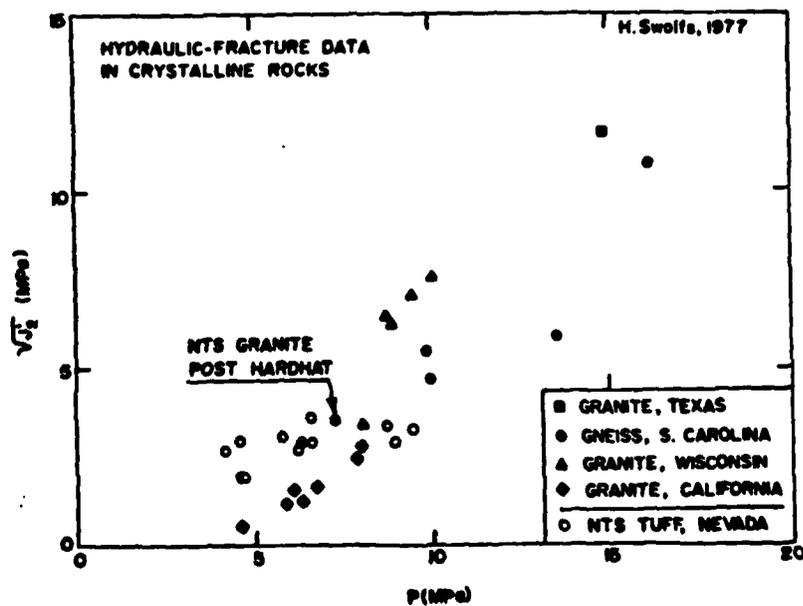


Figure 18. Hydraulic fracture data in crystalline rock. Shear stress invariant vs. mean pressure indicating that these rocks are not under hydrostatic stress in situ.

bars of increased pore pressure to cause the rock to fail (earthquakes). This data has implications of the potential for tectonic stress relief and for "block motion" displacements associated with nuclear events^{25,26}. The equation for horizontal stresses in a near surface region and at depth has been derived (Figure 19) these data will allow us to reasonably predict the stresses for a given geologic environment. It must be emphasized that the "Tectonic environment" must be evaluated in order to assess the relevance of the data. The relationship of stress to age has been compiled in Figure 20 where the ratio of the horizontal minimum stress to the vertical stress is plotted as a function of depth for three groups of rock: granites, gneiss and tuffaceous rock. Note that the younger rocks have a lower stress ratio than the older materials. We may be able to develop a correlation between age and stress ratio, which may enable us to draw conclusions about the *in situ* stress state at various sites based on geotechnical data.

In summary, several conclusions can be drawn: (1) the natural *in situ* stress state is not hydrostatic (2) crystalline rocks show a wide variety of stress ratios while rocks such as salt and shale are nearly under a hydrostatic stress state (3) the existing tectonic environment influences the *in situ* stress state (4) the depth of emplacement of crystalline rock may be important to the *in situ* stress field (5) the *in situ* stress state does have an effect on the seismic radiation pattern and may be a source of energy for "block motion" displacements in fractured rock (6) based on the available data there seems to be a relationship between stress ratio and age of the rock mass.

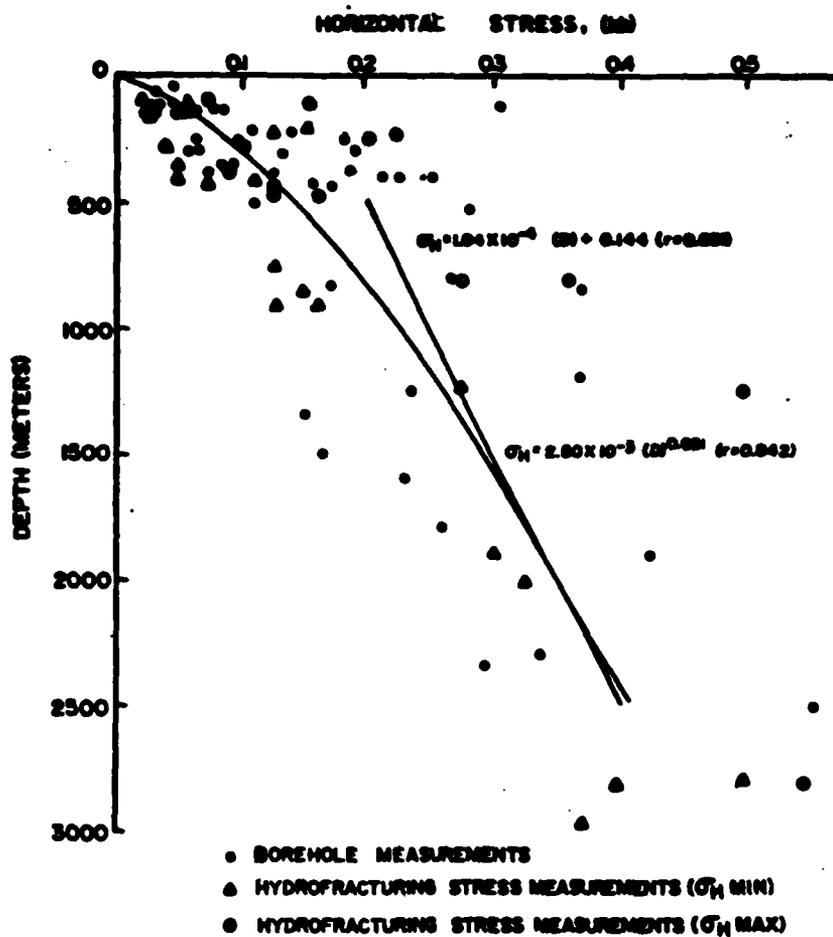


Figure 19. Horizontal stress vs. depth for a variety of stress measurement techniques.

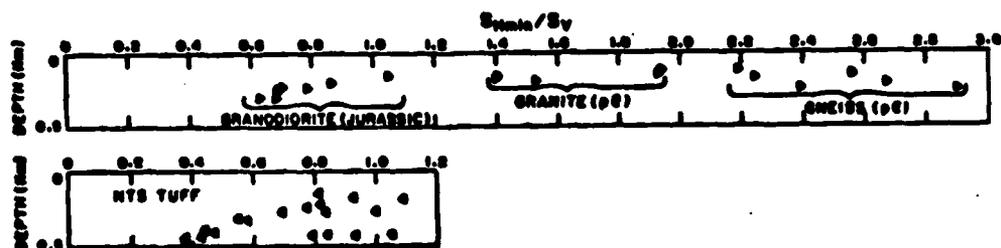


Figure 20. Horizontal minimum/vertical stress ratio as a function of depth for rocks of different ages. Crystalline rocks and tuffaceous rocks are plotted.

3.0 MATERIAL PROPERTY TESTS

3.1 Introduction

This section presents the results of a series of material property tests conducted on a variety of rock types in the support of a calculational program to obtain the reduced displacement potential for a variety of rock types. The properties measured include grain density, bulk density, porosity and degree of saturation. Several kinds of laboratory tests were conducted on three types of rock--sandstone, shale and salt. The tests included uniaxial compression and triaxial tests under a variety of load states. These triaxial load paths include hydrostatic, uniaxial strain, extension and a series of tests conducted to simulate the loads followed during a calculation of the GASSBUGGY site response conducted by PACTECH. In addition, the material properties of a bedded and domal salts were compiled and determined the range of expected physical and mechanical properties.

The GASSBUGGY materials, the Picture Cliffs sandstone and the Lewis shale were chosen because the nuclear event provided some of the best data points for the measurement of reduced displacement potential, and therefore, provided a basis from which the numerical modeling people could determine how well they could calculate the reduced displacement potentials given the laboratory material properties.

3.2 Experimental Procedures

The experimental procedures used for obtaining various physical and mechanical properties are outlined briefly below. (More detailed discussion of experimental procedures, especially for the triaxial test conditions is given in Butters, et al.)²⁹

3.3 GASSBUGGY Event Material Response

The physical and mechanical properties of the two rock types of interest from the GASSBUGGY site, the Picture Cliffs sandstone and the Lewis shale were tested and analyzed in detail. The GASSBUGGY test was a 28 kiloton nuclear event to determine the feasibility of nuclear stimulation of a gas field. The event was detonated in the Lewis shale and had depth of approximately 4,200 feet, approximately 40 feet below the interface between the Lewis shale and the overlying Picture Cliffs sandstone so that both properties of the shale and the sandstone dictate the response of the rock mass of interest (Figure 21)²⁶. A summary of the physical and mechanical properties of the Picture Cliffs sandstone and the Lewis shale are given in Table 6. This is a compilation from both the tests Terra Tek conducted in this program and from other sources.

The response of the Lewis shale to various triaxial loading conditions including compression, uniaxial strain and hydrostatic loading conditions is given in Figures 22-25. Figures 26 through 28 show the same set of tests for the Picture Cliffs sandstone. The Lewis shale shows significant change in the behavior between 1.5 and 2.0 kilobars (Figure 25). At 1.5 kilobars confining pressure of the Lewis shale still fails by brittle fracture; at 2 kilobars, the shale has become ductile and shows no drop in stress at failure. This is typical of a shale unit. In contrast, the Picture Cliffs sandstone (Figure 25), shows a continued brittle failure mode above 2 kilobars confining pressure. Both the sandstone and shale tests were conducted at in situ saturation levels; 2.5% and 1.7% water, respectively, for the sandstone and shale. Failure envelopes for both intact and failed Lewis

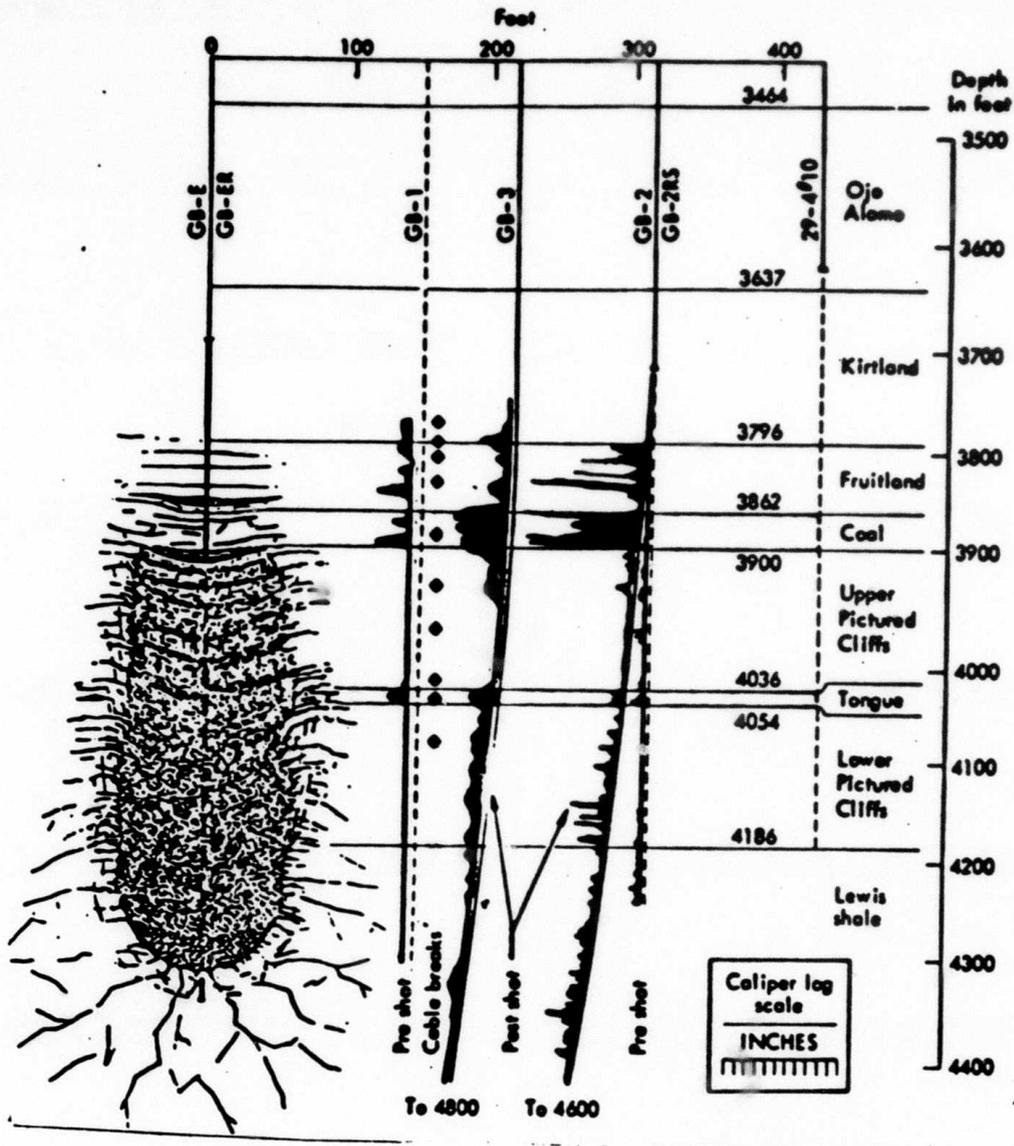


Figure 21. Schematic cross-section through GASBUGGY. The scale of this figure is approximate only. The results of pre-shot and post-shot caliper logs, giving an indication of the fractured nature of the formation are shown.

TABLE 6

PHYSICAL AND MECHANICAL PROPERTIES
OF GASSBUGGY MATERIAL

Property	Picture Cliffs Sandstone	Lewis Shale	
ρ bulk	2.443 ¹	2.586 ¹	gm/cc
ρ grain	2.52 ²	2.62 ²	gm/cc
% water	2.5 ¹	1.7 ¹	
V_p	11,200 ²	13,120 ²	km/s
V_s	8,100 ²	8,040 ²	km/s
ν	.16 ¹	.16 ¹	
k	.10-.16 ²		md

1. Holzer, F. 27

2. Cherry, J. 28

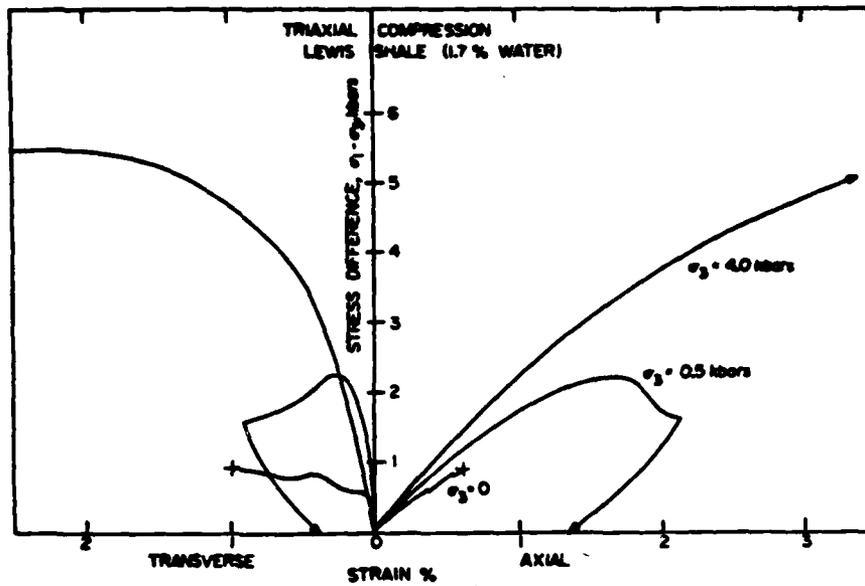


Figure 22. Stress difference vs. axial strain response of the Lewis shale.

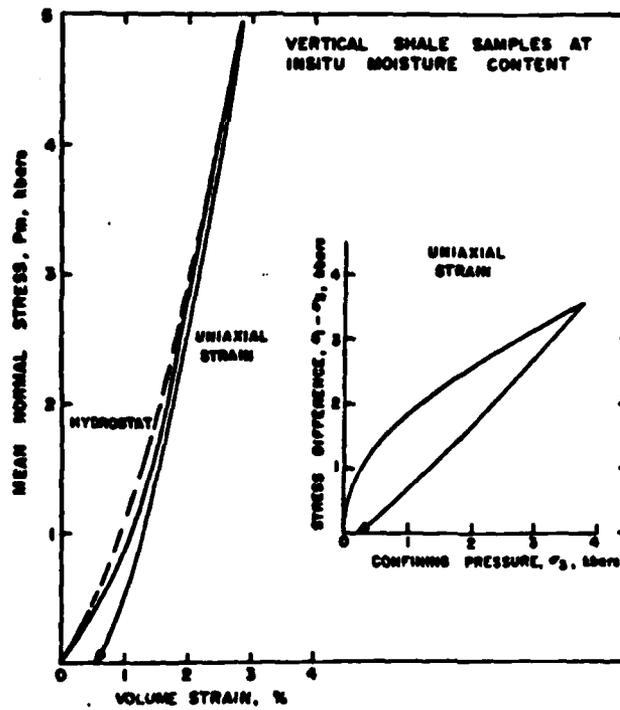


Figure 23. The hydrostatic and uniaxial strain response of Lewis shale. Plots are of volume strain as a function of mean normal stress and stress difference vs. confining pressure.

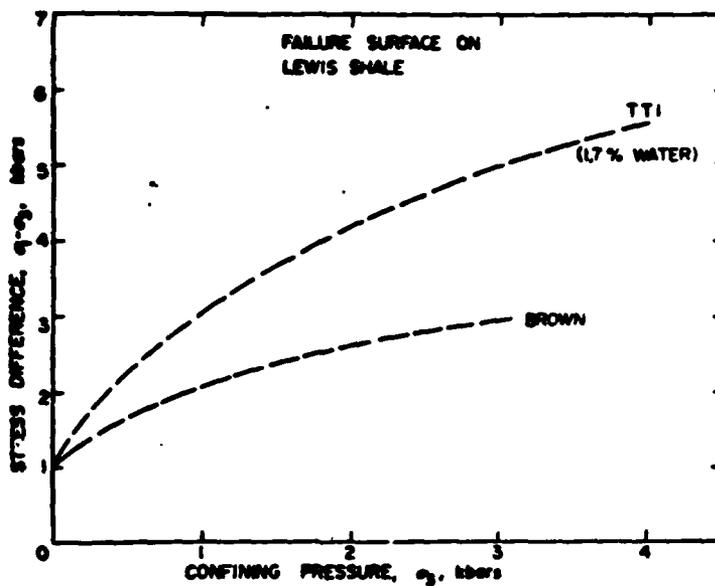


Figure 24. Failure envelope of Lewis shale - Terra Tek (TTi), failure envelope of 1.7% water. Failure envelope from Brown with probably a higher water content.

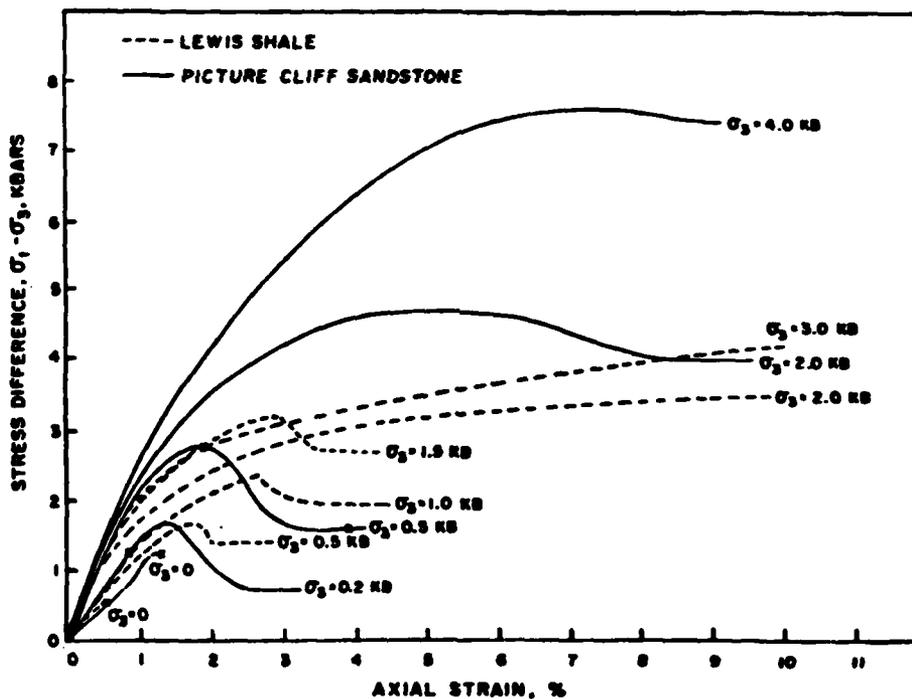


Figure 25. The stress-strain response of Lewis shale and Picture Cliff sandstone up to 4 kilobars confining pressure.

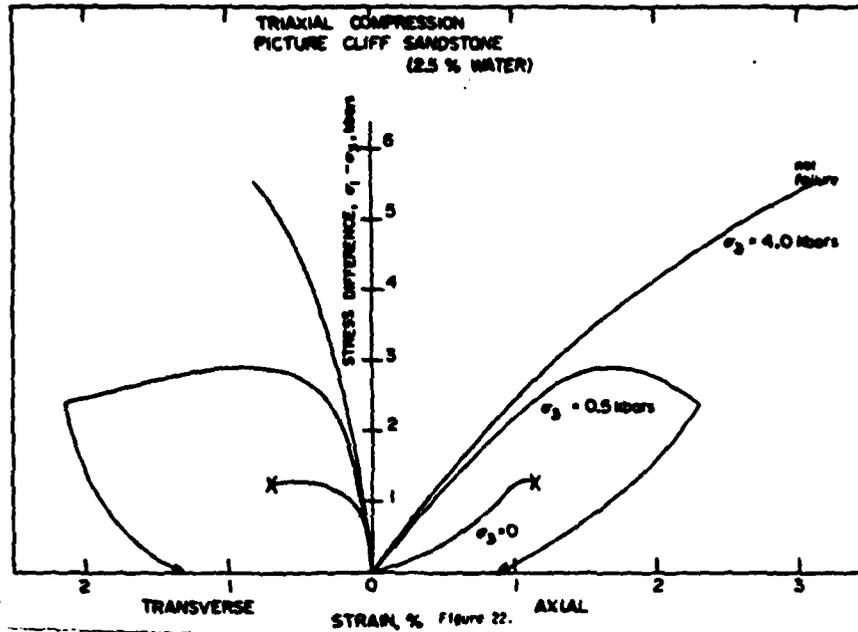


Figure 26. The stress-strain response of the Picture Cliff sandstone up to 4 kilobars confining pressure.

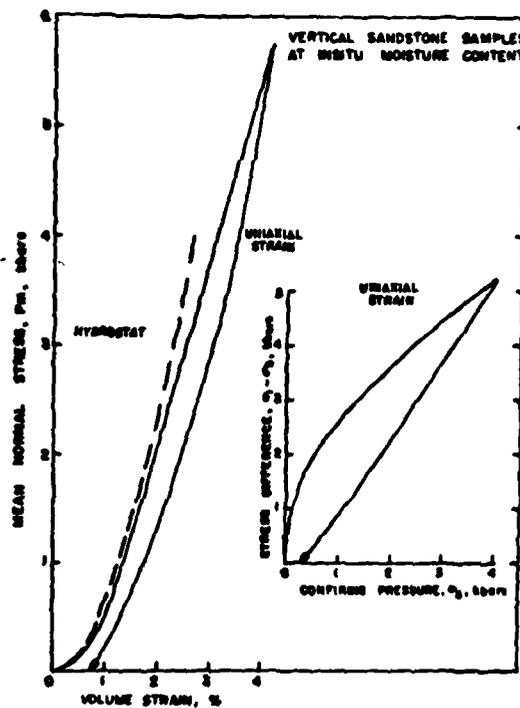


Figure 27. The hydrostatic and uniaxial strain response of Picture Cliff sandstone. The hydrostatic and uniaxial strain response plotted as a function of volume strain vs. mean normal stress and stress difference vs. confining pressure.

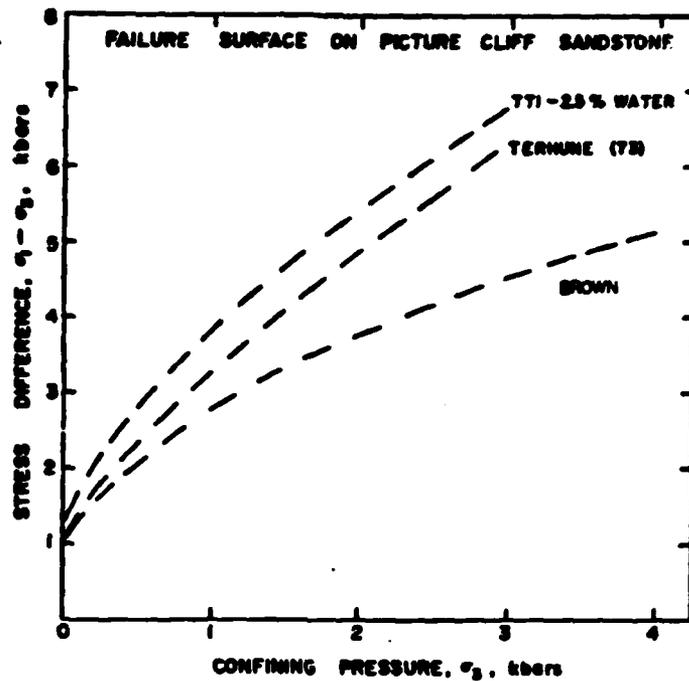


Figure 28. Failure envelope of Picture Cliffs sandstone from three sources of data.

shale and the Picture Cliffs sandstone are shown in Figures 29 and 30. For the Lewis shale and the residual failure envelope approaches that of the intact failure envelope a little over 2 kilobars, but in the Picture Cliffs sandstone difference between the initial and residual failure envelopes remains approximately constant and it would not be expected to intersect until very high pressures where the two curves might merge as shown that for brittle granites. For the Westerly granite, the two failure envelopes do not merge until almost 17 kilobars confining pressure.²⁷ A compilation of the failure envelopes for both the Lewis shale and the Picture Cliffs sandstone indicated that the sandstone is stronger at all confining pressures as would be expected (Figure 31). The shale envelopes fall significantly below the sandstone envelopes especially at confining pressures above 2 kilobars where the shale has become ductile.

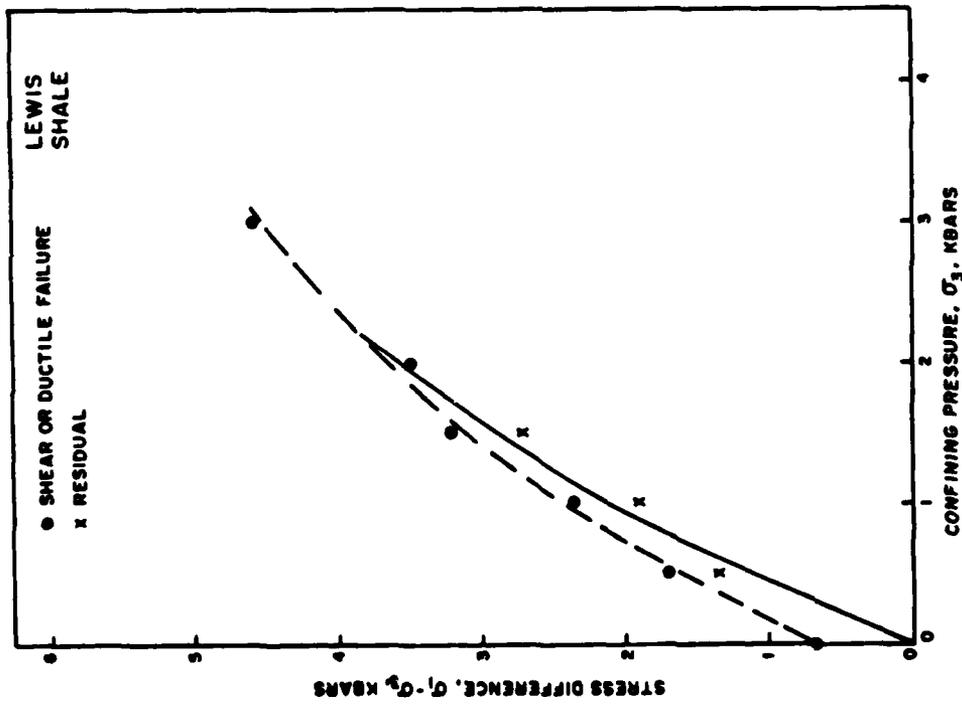


Figure 29. The intact and residual failure envelopes for Lewis shale. Note: the two curves come together at a little over 2 kilobars.

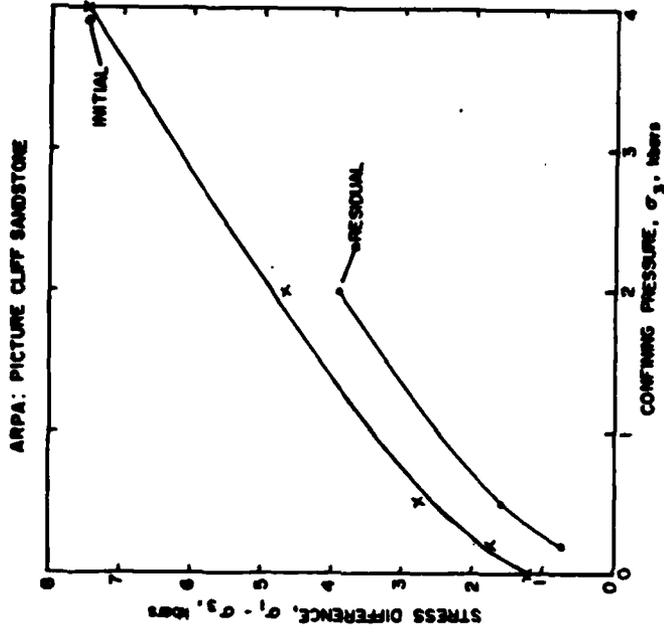


Figure 30. Intact and residual failure envelopes for Picture Cliff sandstone. Note: the two curves are approximately parallel over the range of tests confining pressures studied.

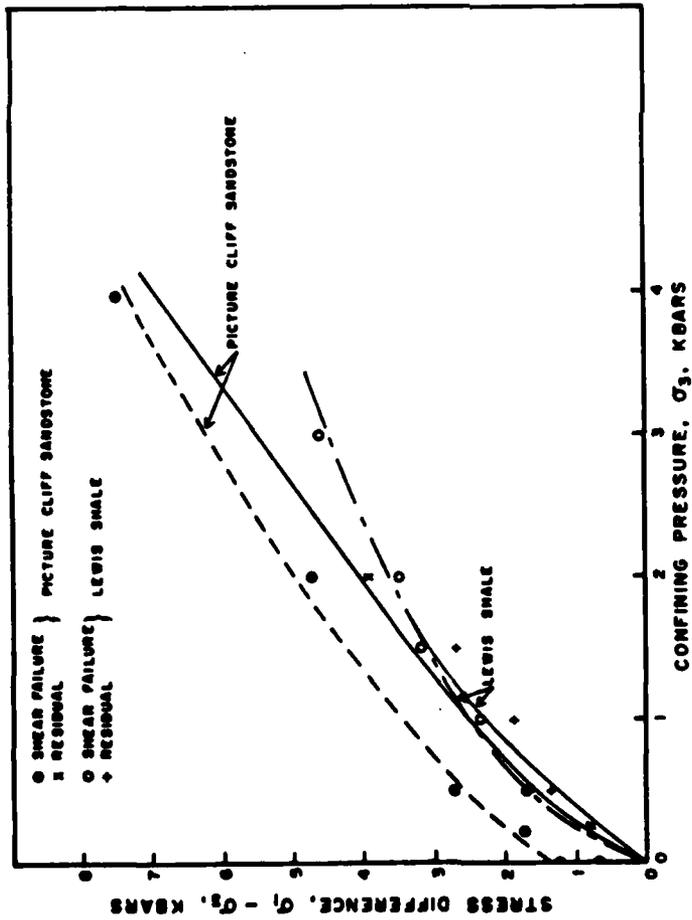


Figure 31. A summary plot of the intact and residual failure envelopes for Lewis shale and Picture Cliff sandstone.

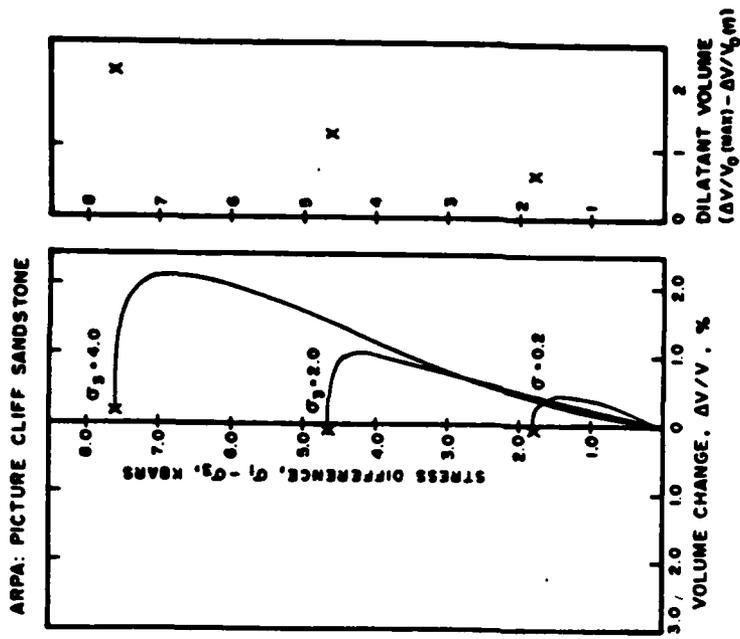


Figure 32. The stress difference vs. volume strain plots illustrating the dilatant volume change occurring in the sandstone.

Dilatancy: The role of dilatancy has important implications on the modeling of material behavior. Dilatant volume represents an available void space prior to failure. This effect coupled with the degree of saturation and permeability will determine the role of pore pressure (effective stress) and the ultimate strength of a material under *in situ* conditions. The plots show the amount of dilatant volume for the sandstone and the shale as a function of stress difference (Figures 32 and 33). As would be expected, the sandstone undergoes an increased dilatancy with confining stress to a maximum of 2.25% at a confining of 4.0 kilobars, a stress difference of 7.1 kilobars (Figure 32). This is significantly more than the dilatant volume for the shale although it is very difficult to pick the failure point of the shale because of its ductile nature beyond the 2 kilobars (Figure 33). In fact, the meaning of dilatant volume for a ductile material such as shale is open to questions since experimentally the failure is continuous shearing without a large stress drop.

Strain Path Effects: A series of tests on both the Lewis shale and Picture Cliff sandstone was conducted along special strain paths dictated by the calculations of Pacifica Technology (PACTECH)²⁹. The results of two sandstone tests are shown as plots of stress difference versus confining pressure and confining pressure versus axial strain (Figures 34 and 35). These tests were conducted following the test procedures outlined previously. The strain paths are labeled in both of the diagrams so that a cross correlation can be made between the two plots.

The sandstone sample was loaded hydrostatically to 0.2 kb and then triaxially to 2.0 kb following prescribed strain paths. The maximum axial strain was 2.75% which equated to a stress difference of 3.5 kilobars.

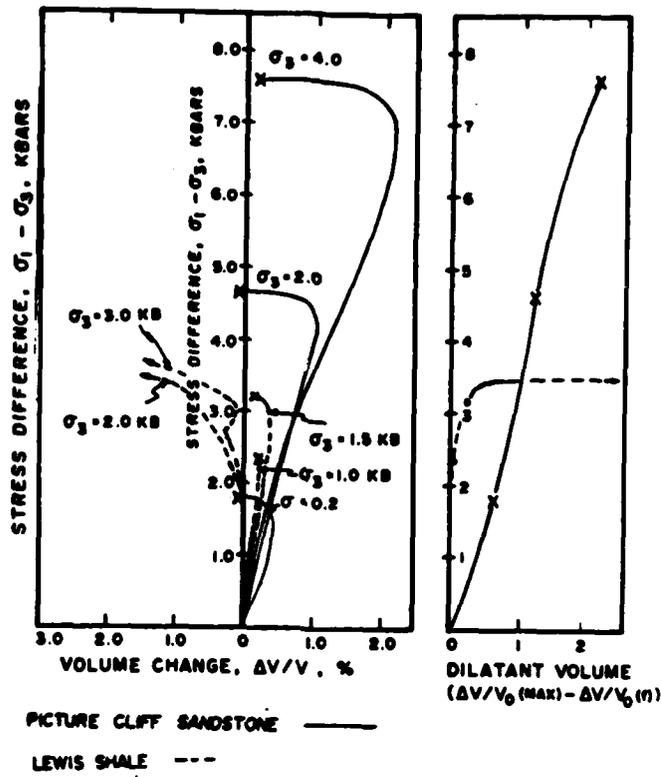


Figure 33. Dilatant volume change as a function of stress difference for the Picture Cliff sandstone and Lewis shale. Note: the difference in the dilatant volume response of the two materials.

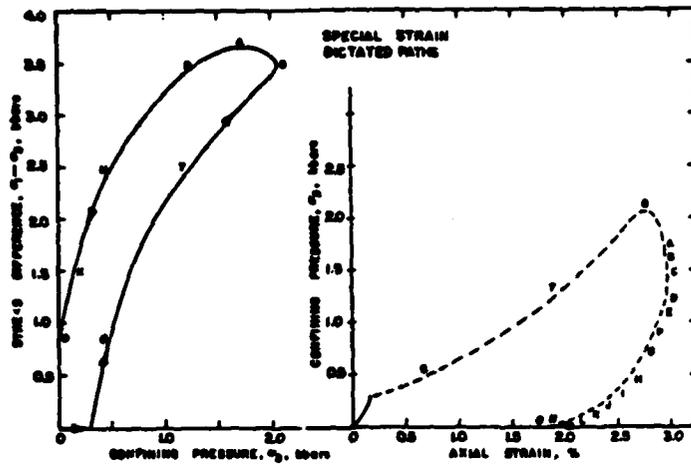


Figure 34. Strain paths loading of Picture Cliff sandstone simulating calculational load path responses by PACTECH.

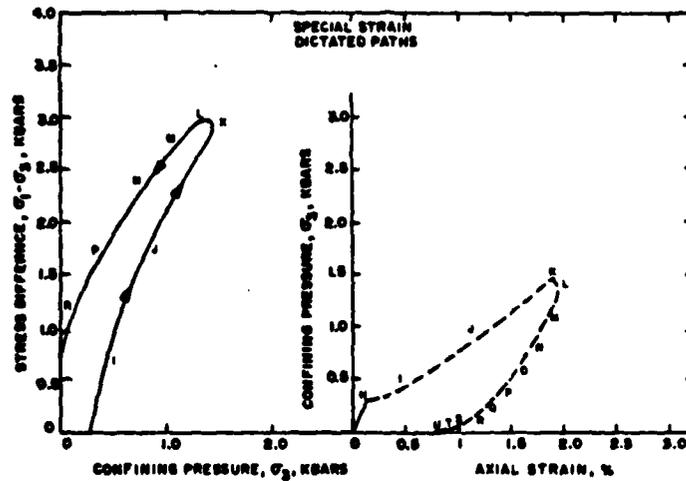


Figure 35. Strain path loading of Picture Cliff sandstone simulating calculational load path responses by PACTECH.

Upon unloading along strain path A-0, the resulting stresses increase to 3.7 kb at A and they decreased along the A-0 path in stress space. Note the stress unloading path is to the left of the loading path indicating that the rock can support a larger stress difference at a lower confining pressure during the unloading cycle as indicated by the strain path. A second set of dictated special strain paths for the Picture Cliffs sandstone conducted a lower confining pressure and resulted in stress differences of only 3 kilobars and maximum axial strain of 2%. Again this test was run by subjecting the sample to the hydrostatic confining pressure to approximately 0.3 of a kilobar and then applying a triaxial loading condition up to about 3 kilobars. The results of these tests were then used to iterate back and forth between Terra Tek and the calculational efforts of Pacifica Technology.

In summary, we looked at and compared several different types of material property phenomena with respect to Lewis shale and the Picture Cliffs sandstone. (1) The Lewis shale becomes ductile above 2 kilobars while the Picture Cliffs sandstone still retains its brittle failure mechanism up to at least 4 kilobars. (2) The residual failure for the shale merges with the intact failure envelope at approximately 2 kilobars while for the sandstone the residual and intact failure envelopes remain essentially parallel up to the highest confining pressures tests (4.0 kilobars). (3) The dilatant behavior of the sandstone is well defined and of course increases with increasing confining pressure. It is more difficult to ascertain the amount of dilatancy prior to failure of the Lewis shale because of the ductile nature of the material above 2 kilobars. However, the shale

undergoes significantly less dilatancy than the sandstone at a given confining pressure until the shale becomes ductile. (4) The compaction under loading of both the sandstone and the shale is quite small due to their low porosity, however under hydrostatic loading the shale is stiffer than the sandstone although it is not able to support as high a differential stress at a given confining pressure as the sandstone.

Strain Rate Effects on Modulus: The effect of strain rate on the dynamic modulus of rock may be important to the development of a realistic constitutive model for rock masses³⁰. Strain rate effects may be seen under conditions of stress relaxation under constant strain and creep under constant. To analyze these effects we (1) developed an analytical model of a viscoelastic standard solid (2) developed a test technique and experimental apparatus to measure the relaxation modulus (E_t). The model development is given in detail in Appendix A. An experimental apparatus was developed to load a specimen in a 1.-10 millisecond rise time to a few bars by dropping a suspended weight on to the specimen and analyzing the stress relaxation with a laser interferometer. A schematic of the experimental is given in Figure 36 and a photograph of the apparatus in Figure 37.

Representative data from tests on Picture Cliffs sandstone are given in Figures 38 and 39 respectively. The data for the sandstone shows a rise time of 6 milliseconds to a stress level of 9.3 bars and for the shale, a rise time of 2.3 milliseconds to $\sigma = 9.3$ bars. From the model the relaxation modulus was calculated for the sandstone as follows:

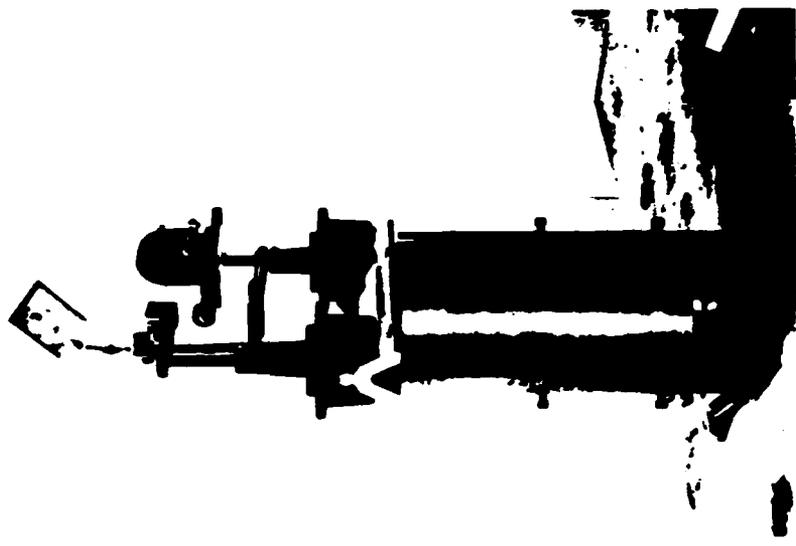


Figure 37. Photograph of experimental apparatus with tests specimens.

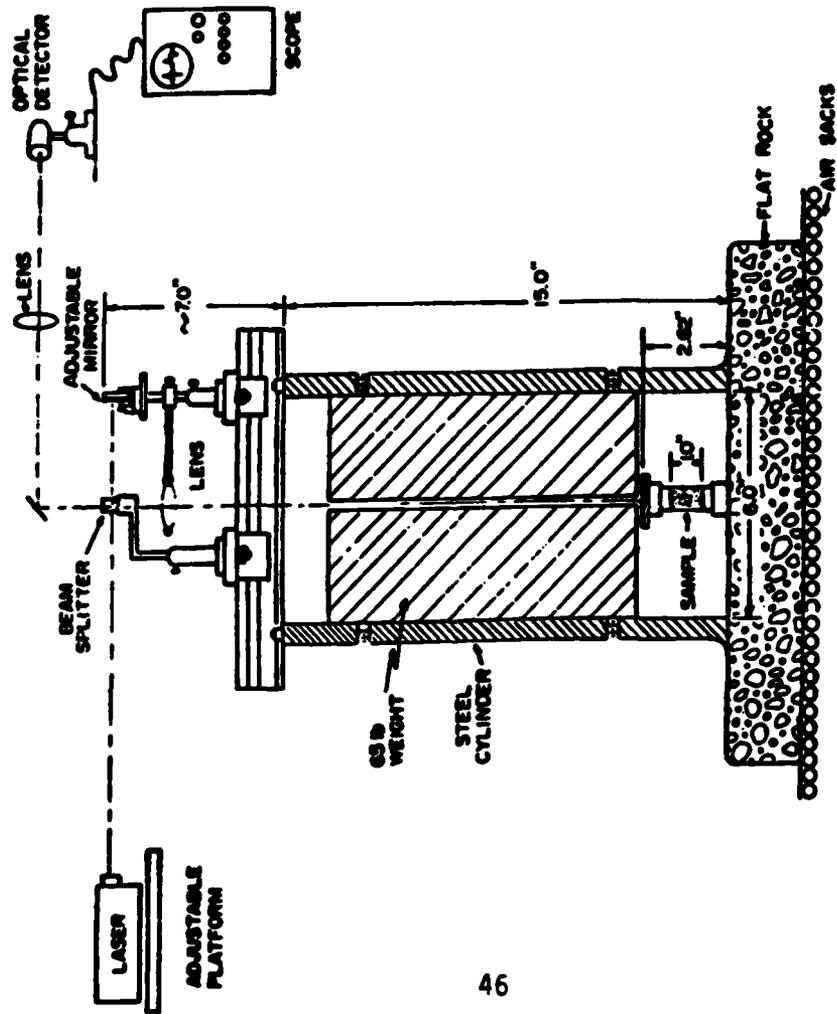


Figure 36. Schematic diagram of the experimental apparatus for the measurement of low stress level dynamic modulus.

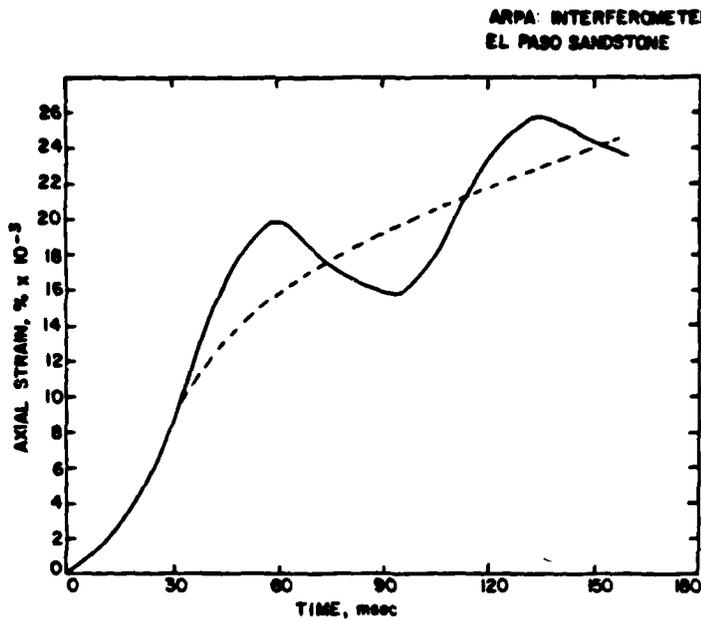


Figure 38. Axial strain vs. time for the low stress level dynamic tests on Picture Cliff sandstone.

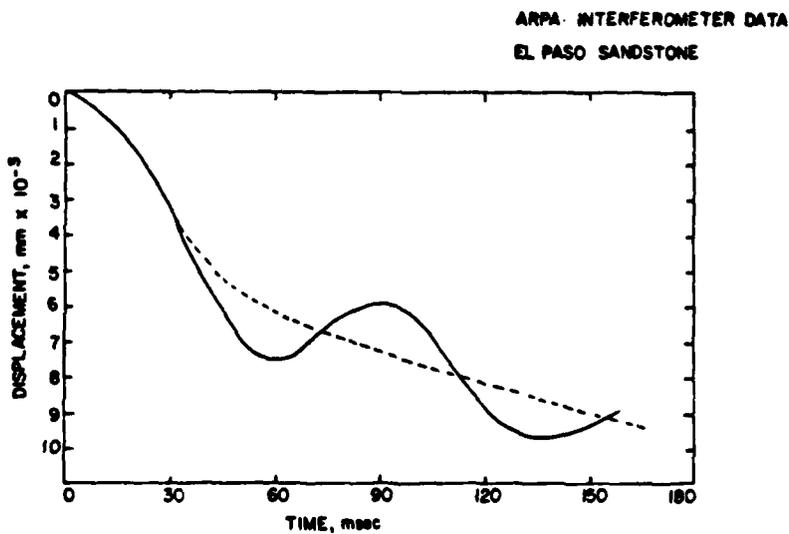


Figure 39. Displacement vs. time for the low stress level dynamic tests conducted upon the Picture Cliff sandstone.

From the experiment:

$$\begin{aligned}t_0 &= 6 \text{ msec} \\ \tan \alpha &= 10 \text{ } \mu\text{s/msec} \\ \epsilon(\infty) &= 80 \text{ } \mu\text{s} \\ \epsilon(t_0) &= 63 \text{ } \mu\text{s} \\ \sigma_0 &= 135 \text{ lb/m}^2\end{aligned}$$

Then:

$$\tan \alpha = \frac{\sigma_0}{t_0 R_1}$$

$$\epsilon(\infty) = \sigma_0 \left(\frac{1}{R_1} + \frac{1}{R_2} \right)$$

$$\epsilon(t_0) = \frac{\sigma_0}{t_0} \left(\frac{t_0}{R_1} + \frac{t_0}{R_2} - \frac{\eta}{R_2} + \frac{\eta}{R_2} e^{-\frac{R_2}{\eta} t_0} \right)$$

$$R_1 = \frac{\sigma_0}{t_0} \tan \alpha = 2.25 \times 10^6 \text{ lbs/in}^2$$

$$R_2 = \frac{\sigma_0}{\epsilon(\infty) - \frac{\sigma_0}{R_1}} = 6.75 \times 10^6 \text{ lb/in}^2$$

Solving (3) for η gives:

$$0 = 3.469 \times 10^7 - \eta + \eta e^{-\frac{40.5 \times 10^6}{\eta}}$$

$$\eta = 1.27 \times 10^8$$

Relaxation Modulus

$$E(t) = 1.687 \times 10^6 (1 - e^{-0.07t}) + 2.25 \times 10^6 e^{-0.07t}$$

$$E(t) = 5.63 \times 10^5 e^{-0.07t} + 1.687 \times 10^6$$

Using the following values for the shale:

$$t_0 = 2.3 \text{ msec}$$

$$\tan \alpha = 8 \text{ } \mu\text{s/msec}$$

$$\epsilon(t_0) = 18.5 \text{ } \mu\text{s}$$

$$\epsilon(\infty) = 23 \text{ } \mu\text{s}$$

$$\sigma_0 = 135 \text{ lb/in}^2$$

$$E(t) = 1.46 \times 10^6 e^{-0.02t} + 5.87 \times 10^6$$

A comparison of the moduli from static, impact and ultrasonic (dynamic) tests are summarized in Table 7. For both the Lewis shale and Picture Cliff sandstone the static moduli was considerably less than the dynamic moduli. In the case of the Picture Cliff sandstone it was a factor of 2 less (76 vs. 152 kb) than impact test results and almost a factor of three less than moduli calculated from ultrasonic tests. The data from the Lewis shale also indicates a significant difference between static and dynamic moduli. However, the impact tests gave a higher value than the ultrasonic data for the Lewis shale. Aluminum was used as a calibration sample for comparison with published data.

TABLE 7

Modulus Comparisons

<u>Material</u>	<u>Test Type</u>	<u>Young's Mod (kbars)</u>
Picture Cliff Sandstone (dry)	Static	76
	Impact	152
	Ultrasonic	221
Lewis Shale (dry)	Static	159 for stress < 500 psi
		338 for stress > 500 psi
	Impact	503
	Ultrasonic	400
Aluminum	Impact	690

In summary, the experimental technique proved moderately successful and values of relaxation modulus were obtained in the loading range 2-6 msec. Improvement in the loading mechanism is required so that ringing does not occur. In addition more rapid loading in a 1.0 msec. risetime would be beneficial as would a method to vary the loading rate equivalent to a frequency range of 5-100 Hz. in order to better evaluate rate effects.

3.4 Salt

The physical and mechanical properties of bedded salt from the GNOME event and dome salt from the SALMON event were obtained. The physical properties are listed in Table 8. These properties can be compared to data on a variety of salt types (bedded, domal, polycrystalline, single crystal) obtained from the literature (Table 9).

The mechanical properties were obtained under both triaxial compression and extension. Plots of axial stress at various confining pressures indicates that salt becomes ductile at approximately 1.5% strain even at very low confining pressures (250 psi) (Figure 40). Triaxial tests up to a confining pressure of 4 kilobars indicates that both bedded and dome salt are "ductile" at a confining pressure of 0.75 kb and reached maximum stress difference of 0.68 kb (Figure 41). This is slightly less strength (12%) than the 0.8 kb measured for the ultimate strength of polycrystalline salt measured previously. Heard's data also indicated that ultimate strength was independent of strain above 2% strain (Figure 42) for polycrystalline salt³⁶.

TABLE 8
PROPERTIES OF GNOME AND SALMON SALTS

	GNOME ¹	GNOME ²	SALMON ³	SALMON ⁴
ρ_0 (gm/c)	2.13-2.46	2.136	1.89	2.214
ρ_{grain} (gm/cc)	---	2.173	----	2.236
Porosity	.76-5.1	1.730	----	1.000
% water	1.00	.026	----	.019
V_p (km/sec)	4.08		4.30	
V_s (km/sec)	2.88		2.35	
K (kbar)			2.31	
G (kbar)			1.20	
ν	0.28-0.32		0.26	
Depth (m)		309.45		790.55

¹Heart (1963)³¹, ²Pratt, et al. (1978), ³Christensen (1964)³²

⁴Pratt, et al. (1978)

TABLE 9

REPRESENTATIVE SALT PROPERTIES

PROPERTY	UNITS	DOME	BEDDED	POLY XTAL.	SINGLE XTAL.
ρ bulk	gm/cm ³	2.16 ³ 2.17 ⁶	2.17(w/clay) ⁷ 2.17(w/anh) ⁷	2.14 ⁴	
porosity	%		0.5 ⁷	1.11 ⁴ 4.16 ⁸	
ν_{ult}	kbars	0.38 ¹ 0.14 ² 0.20 ² 0.21 ³ 0.27	0.27±.03(Δ) ¹ 0.29±.03(Π) ¹ 0.16±.03(w/clay) ⁷ 0.25±.04(w/anh) ⁷ 0.44(3=0.03) ⁶	0.56 ⁴	0.26(1 samp) ⁵
T_0	kbars	0.008 ³	0.014±.003(w/clay) ⁷ 0.016±.001(w/anh) ⁷	0.038 ⁴	0.42(2 samp) ⁵
ν_p	km/s	4.1 ³		4.1 ⁴	
ν_s	km/s	4.2(.051kb) ³		4.4 ⁴ (0.48kb) ⁴ 2.4 ⁴ 2.7(0.48kb) ⁴	
E_t	kbars	39±3 ¹ 54 to 120 ² 95 to 160 ² 110(10% ult) ³	24±12 (Δ) ¹ 2±4 (Π) ¹ 140±40(w/clay) ⁷	120(initial) ⁸	
ν (shear)	kbars			136(initial) ⁴ 122(dynamic) ⁴ initial	
ν	-	0.19(initial) ³ 0.35(10% ult) ³	0.18-0.27 ⁷	0.28(static) ⁴ initial	
σ yield pt.	kbars	0.20 ¹ 0.08 ² 0.12 ² 0.10 ³	0.16±.02 (Δ) ¹ 0.17±.03 (Π) ¹	0.23(dynamic) ⁴ initial	0.075 ⁵
k(bulk)	kbars	37-56(int) ² 58(initial) ²		211(triaxial) ⁴ initial 109(hydrostatic) initial 184(dynamic) ⁴ initial	238(hyd.int) ⁴

1 Pariseau, W. 33

2 Serrata, S. 34

3 Deere, D. 35

4 Heard, H. 36

5 Handin, J. 37

6 Hansen, R. 38

7 Wawersik 39

8 Dropek, R. 40

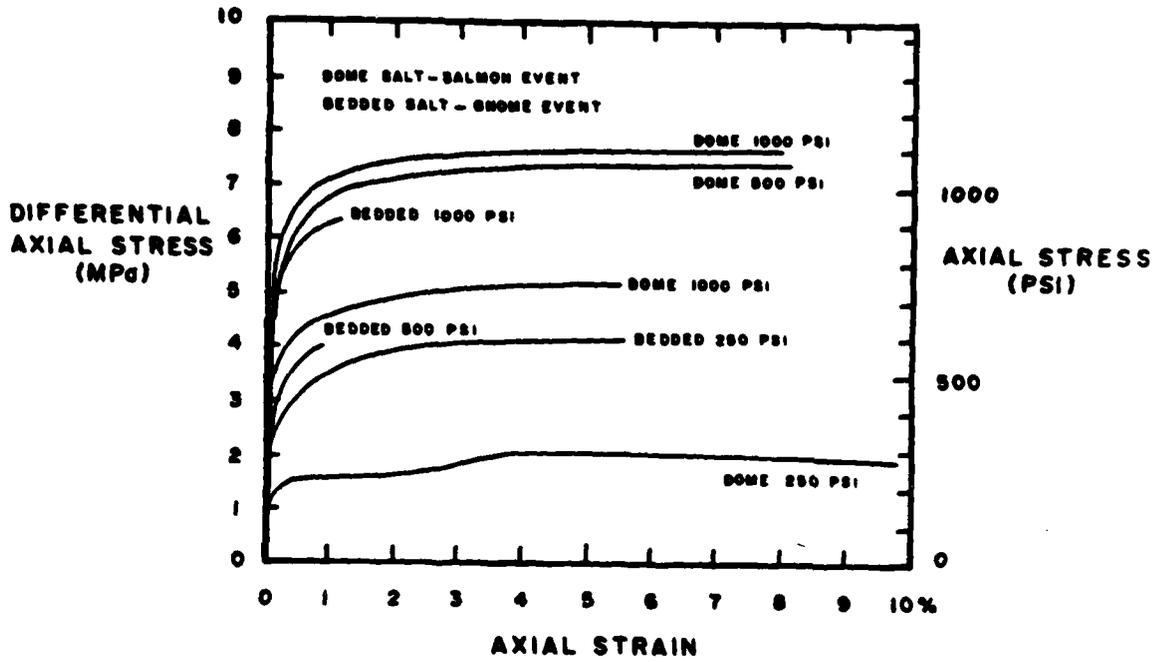


Figure 40. Stress-strain response of bedded and dome salt at various confining pressures.

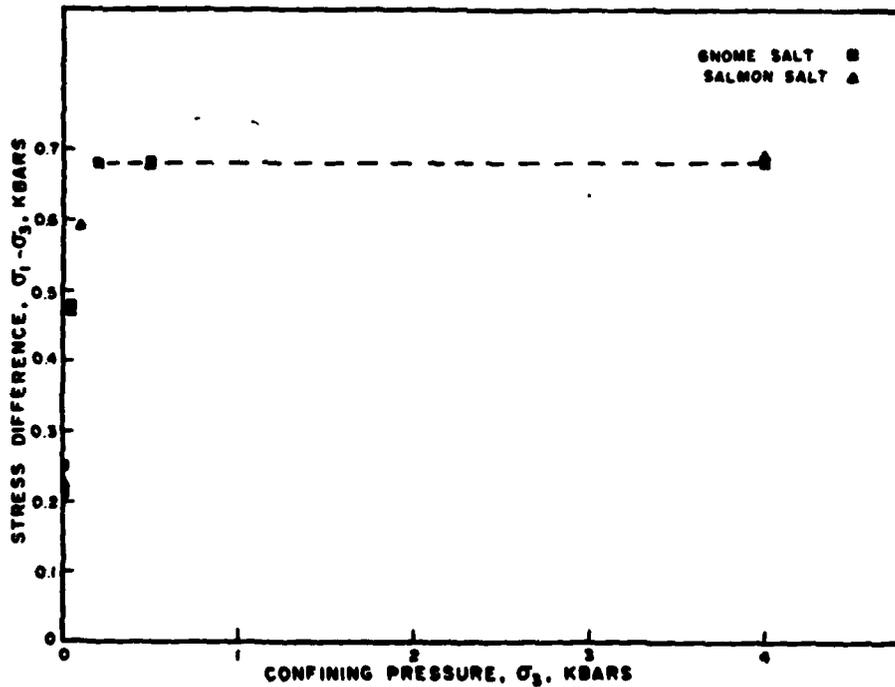


Figure 41. Failure envelopes for GNOME (bedded salt) and SALMON (dome salt) as a function of confining pressure up to 4 kilobars.

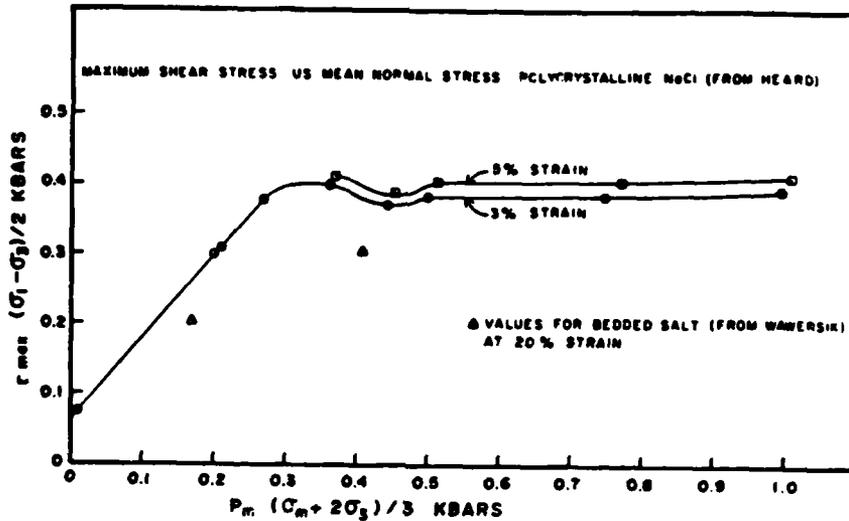


Figure 42. Failure envelope of polycrystalline salt plotted as maximum shear stress vs. mean normal stress at constant strain values³⁶.

Cylindrical samples nominally 1 inch (25 mm) in diameter by 1.5 inch (37 mm) in length were tested in extension with confining pressure of 250, 500 and 1,000 psi (1.72, 3.45 and 6.89 MPa respectively). The specimen ends were surface ground flat and parallel to within ± 0.01 mm and the specimens were jacketed with a double layer of polyurethane, .01 inch (.25 mm) thick. Steel end caps were attached to the specimens with 5 minute epoxy and axial strain cantilevers were attached to the end caps with set screws. A hook was attached to the bottom end cap and a universal joint and loading piston were attached to the top.

The specimen assembly was then lowered into a small pressure vessel and rotated 90° to engage the hook into its seat in the bottom of the vessel. The upper loading piston provided the seal at the top of the vessel by engaging 2 O-rings.

The pressure required to overcome the seal friction was measured to be about 200-250 psi (1.4-1.7 MPa) over the range of confining pressure used. The hydraulic ram from the testing machine frame was then lowered to touch the top of the loading piston and the confining pressure raised to the desired value. Then the ram was raised at 1.5×10^{-3} in/sec, allowing extension of the specimen at the nominal strain rate of 10^{-3} sec^{-1} until failure.

The combinations of the seal friction and electronic noise resulting from the extremely low differential axial load required for extension of such a small specimen resulted in an overall uncertainty in the differential axial stress measurement of about 250 psi (1.72 MPa).

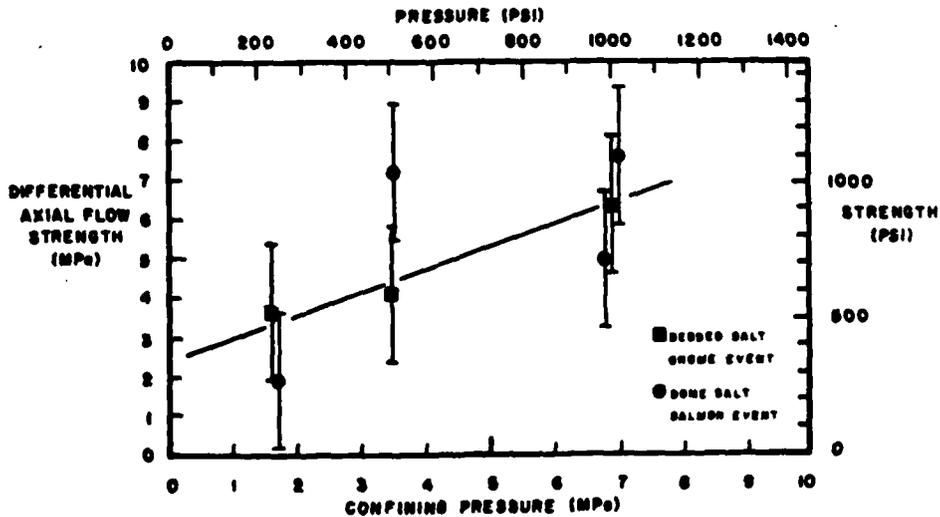


Figure 43. Extension failure envelopes. Differential axial strength as a function of confining pressure for both bedded and domal salt.

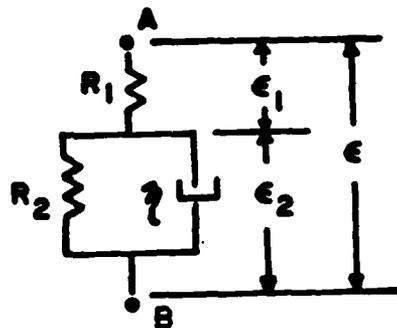
The extension failure envelopes (Figures 43) shows an increase in strength with confining pressure up to 7 MPa (70 bars). The average strength is approximately 6.2 MPa (62 bars) at 7.0 MPa confining stress. The data scatter is significant due to the nature of salt. With the wide scatter in the data as indicated by the bars, there is no indication that one kind of salt (bedded vs. dome) is stronger in extension than the other.

APPENDIX A

STRESS-STRAIN RESPONSE OF VISCOELASTIC STANDARD SOLID MODEL

Standard Solid Viscoelastic Model

Strain rate effects seen in rocks when stresses are applied are manifested by stress relaxation under constant strain and creep under constant stress. Mathematically, this response can be modeled by combining various systems of linear springs and viscous dash pots. One such model is the standard solid model as shown below in Figure A.



- R - Elastic spring constant
- η - Co-efficient of viscosity
- ϵ - Strain

The standard solid model is simply a linear spring in series with a Kelvin model (spring and dash plot in parallel). Constitutive equations for this viscoelastic model are developed as follows:

When a stress is applied across the sample at points A and B, time dependent strain results. Total strain at any time (t) is the sum of the individual element strains.

$$\epsilon = \epsilon_1 + \epsilon_2 \quad (1)$$

where

$$\text{For the spring} \quad \epsilon_1 = \frac{\sigma}{R_1} \quad (2)$$

And for the Kelvin element

$$\dot{\epsilon}_2 + (R_2/\eta) \epsilon_2 = \frac{\sigma}{\eta} \quad (3)$$

In the LaPlace space

$$\hat{\epsilon}_1 = \frac{\hat{\sigma}}{R_1} \quad (4)$$

$$\hat{\epsilon}_2 = \frac{\hat{\sigma}}{\eta(S + \frac{R_2}{\eta})} \quad (5)$$

so

$$\hat{\epsilon} = \frac{\hat{\sigma}}{R_1} + \frac{\hat{\sigma}}{\eta(S + \frac{R_2}{\eta})} \quad (6)$$

Rearranging

$$\frac{\eta}{R_2} S \hat{\epsilon} + \hat{\epsilon} = \frac{\hat{\sigma}}{R_1} + \frac{\eta}{R_1 R_2} S \hat{\sigma} + \frac{\hat{\sigma}}{R_2} \quad (7)$$

Taking the inverse LaPlace

$$\frac{\eta}{R_2} \dot{\epsilon} + \epsilon = \frac{R_1 R_2}{R_1 R_2} \sigma + \frac{\eta}{R_1 R_2} \dot{\sigma} \quad (8)$$

Rearranging to general differential equation form

$$\sigma + \frac{\eta}{R_1 + R_2} \dot{\sigma} = \frac{R_1 R_2}{R_1 + R_2} \epsilon + \frac{R_1 \eta}{R_1 + R_2} \dot{\epsilon} \quad (9)$$

Relaxation Modulus

In a relaxation test a step of constant strain $\epsilon = \epsilon_0 H(t)$ is applied and the stress $\sigma(t)$ is measured as shown in Figures B and C. If the material behavior is linear, the stress can be represented by:

$$\sigma(t) = \epsilon_0 E(t) \quad (10)$$

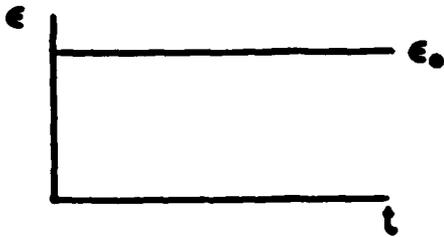


Figure B

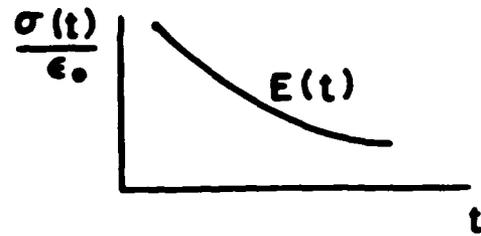


Figure C

The function $E(t)$ thus obtained is called the relaxation modulus. It is the stress per unit of applied strain and is different for each material.

Inserting a unit step of strain into the standard solid differential equation (9) yields

$$\sigma + \frac{\eta}{R_1 + R_2} \dot{\sigma} = \frac{R_1 R_2}{R_1 + R_2} \epsilon_0 H(t) + \frac{R_1 \eta}{R_1 + R_2} \epsilon_0 \delta(t) \quad (11)$$

In the LaPlace space

$$\sigma + \frac{\eta}{R_1 + R_2} (S) \hat{\sigma} = \frac{R_1 R_2}{R_1 + R_2} \left(\frac{1}{S}\right) \epsilon_0 + \frac{R_1 \eta}{R_1 + R_2} \epsilon_0 \quad (12)$$

or

$$\hat{\sigma} = \left[\frac{\frac{R_1 R_2}{R_1 + R_2} \left(\frac{1}{S}\right) + \frac{R_1 \eta}{R_1 + R_2}}{1 + \frac{\eta}{R_1 + R_2} S} \right] \epsilon_0 \quad (13)$$

Taking the inverse LaPlace and arranging terms

$$\sigma(t) = \left[\frac{R_1 R_2}{R_1 + R_2} + \left(R_1 - \frac{R_1 R_2}{R_1 + R_2} \right) e^{-\left(\frac{R_1 + R_2}{\eta}\right) t} \right] \epsilon_0 \quad (14)$$

The relaxation modulus is therefore

$$E(t) = \left[\frac{R_1 R_2}{R_1 + R_2} + \left(R_1 - \frac{R_1 R_2}{R_1 + R_2} \right) e^{-\left(\frac{R_1 + R_2}{\eta}\right) t} \right] \quad (15)$$

Experimental Determination of R_1 , R_2 and η

To determine a numerical value for the relaxation modulus, the spring constants and coefficient of viscosity must be known. These values can be obtained from a creep or relaxation test. The test chosen was an impact test where a ramp stress followed by constant stress was applied and the resulting strain was measured. The test is represented in Figures D and E.

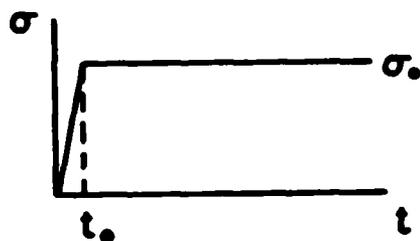


Figure D

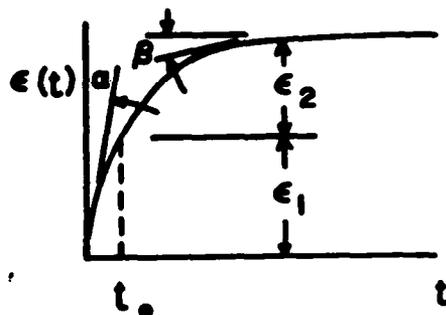


Figure E

From this test, one can measure α , β , $\epsilon(t)$ and $\dot{\epsilon}(t)$. Measurement of these quantities allows determination of R_1 , R_2 and η as shown by the exact differential equations for this type loading.

For $t < t_0$

$$\sigma = \frac{\sigma_0}{t_0} t \quad (16)$$

$$\dot{\epsilon} = \frac{\sigma_0}{t_0} \left(\frac{1}{S^2} \right) \quad (17)$$

Inserting into equation (6)

$$\dot{\epsilon} = \left[\frac{1}{R_1} + \frac{1}{\eta \frac{R_2}{\eta} + S} \right] \frac{\sigma_0}{t_0} \left(\frac{1}{S^2} \right) \quad (18)$$

Inverting from the LaPlace

$$c(t) = \frac{\sigma_0}{t_0} \left[\frac{t}{R_1} + \frac{t}{R_2} - \frac{\eta}{R_2^2} + \frac{\eta}{R_2^2} e^{-\frac{R_2 t}{\eta}} \right] \quad (19)$$

Differentiating

$$\dot{c}(t) = \left[\frac{\sigma_0}{t_0} \left(\frac{1}{R_1} + \frac{1}{R_2} - \frac{1}{R_2} e^{-R_2/\eta t} \right) \right] \quad (20)$$

at $t = 0^+$

$$\dot{c}(0^+) = \frac{\sigma_0}{t_0 R_1} = \tan \alpha \quad (21)$$

For $t > t_0$

$$\alpha = \frac{\sigma_0}{t_0} t - \frac{\sigma_0}{t_0} (t - t_0) \quad (22)$$

$$\hat{\alpha} = \frac{\sigma_0}{t_0} \left(\frac{1}{S^2} \right) - \frac{\sigma_0}{t_0} \left(\frac{1}{S^2} \right) e^{-st_0} \quad (23)$$

Inserting into equation (6)

$$\hat{c} = \left[\frac{1}{R_1} + \frac{1}{\eta(R_2 + S)} \right] \left[\frac{\sigma_0}{t_0} \left(\frac{1}{S^2} \right) - \frac{\sigma_0}{t_0} \left(\frac{1}{S^2} \right) e^{-st_0} \right] \quad (24)$$

Inverting from the LaPlace

$$c(t) = \frac{\sigma_0}{t_0} \left[\frac{t_0}{R_1} + \frac{t_0}{R_2} + \frac{\eta}{R_2^2} e^{-R_2/\eta t} (1 - e^{-R_2/\eta t_0}) \right] \quad (25)$$

At $t = \infty$

$$c(\infty) = \sigma_0 \left[\frac{1}{R_1} + \frac{1}{R_2} \right] \quad (26)$$

Differentiating

$$\dot{c}(t) = \frac{1}{R_2} e^{-R_2/\eta t} (1 - e^{-R_2/\eta t_0}) \quad (27)$$

At $t = \infty$

$$\dot{\epsilon}(\infty) = 0 = \tan \beta \quad (28)$$

Thus, by substituting measured values of α , $\epsilon(t_0)$ and $\epsilon(\infty)$ into equations 19, 21 and 26, one can calculate numerical values for R_1 , R_2 and η . These values can then be inserted into the relaxation modulus.

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20. ABSTRACT (Continue on reverse side if necessary and identify by block number) This report includes a compilation of both physical and mechanical property data for use in developing in situ constitutive relations. The in situ stress state is defined for a variety of rock types and structural environments. Triaxial tests, including hydrostatic and uniaxial strain tests, were conducted on sandstone and shale from the GASBUGGY site. Problems studied include the intact versus residual failure envelopes, dilatant behavior and the dependency of modulus on strain-rate. Salt cores from both the GNOME event and SALMON event were tested under triaxial conditions; both compression and extension failure envelopes were determined. Originator Keywords		