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A STUDY OF GEOLOGIC HAZARDS AND GEOTECHNICAL INPUT FOR SELECTED CRITICAL FACILITIES - CACHE VALLEY, UTAH

by

Kenneth Robert Green

A thesis submitted in partial fulfillment of the requirements for the degree

of

MASTER OF SCIENCE

in

Engineering

Approved:

UTAH STATE UNIVERSITY Logan, Utah

ACKNOWLEDGMENTS

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The Utah Geological and Mineral Survey initiated this study under the direction of Bruce N. Kaliser in furtherance of the objectives of the Geological Survey's earthquake hazard reduction program. The field data was collected by the writer while temporarily employed by the Utah Geological and Mineral Survey. The writer very gratefully acknowledges the help of Mr. Kaliser.

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ABSTRACT

A Study of Geologic Hazards and Geotechnical Input for Selected Critical Facilities - Cache Valley, Utah

by

Kenneth Robert Green, Master of Science Utah State University, 1977

Major Professor: Dr. Loren Runar Anderson Department: Civil and Environmental Engineering

Important public facilities located in Cache Valley, Utah, were studied to assess their exposure to geologic hazards. Also, the level of geotechnical and geologic expertise involved in the siting, design, and construction of critical public facilities was studied. A discussion of several individual facilities, that were exposed to some degree of hazard, was presented to illustrate the nature of the problem.

Earthquake hazards probably constitute the greatest threat to facilities and to the general welfare of the public. Many earthquake hazard mitigation measures were discussed and recommendations were made, based on aseismic design criteria which has been developed for seismically active areas such as California.

A goal of this study was to present information concerning the seriousness of the current geologic hazard situation in Cache Valley, Utah, in an attempt to help generate public interest in understanding and correcting the problem.

CHAPTER 1

INTRODUCTION

Nature of Problem

Cache Valley and other parts of Utah are susceptible to certain types of geologic hazards. For the purpose of this study a geologic hazard has been defined as a geologic condition that poses a threat to a structure or community in a manner that threatens life, property, and essential activities. This study was conducted to evaluate geotechnical input and to identify geologic hazards that could influence vital facilities used by the public.

Certain geologic processes or events pose hazards which if not properly guarded against could cause damage or destruction. This damage could in turn lead to social and political disorder, loss of life, and loss of financial or physical investment. Proper planning, analysis, and design of structures and appropriate consideration of the natural forces acting on them will minimize the negative effects of disruptive forces.

During the course of this study, lifeline facilities, emergency service structures, and other important public facilities were studied from a geotechnical and geologic viewpoint. Lifeline facilities are utilities such as pipelines, transmission lines, transportation and communication lines which are owned and operated by public entities and utility companies. These facilities are classified into four general categories by the American Society of Civil Engineers and are listed below:

- Energy
- Water
- Transportation
- Communication

A further breakdown of lifelines is shown in Table 1-1 after Duke (1975).

Table 1-1. Lifeline facilities (after Duke, 1975).

Energy	Water
Electricity	Potable
Gas	Flood
Liquid fuel	Sewage and solid waste
Transportation	Communication
Highway	Telphone and telegraph
Railway	Radio and television
Airport	Main and press
Harbor	

Emergency service structures can be identified as:

- Government buildings housing principal emergency operating centers
- Police stations and security cells
- Fire stations and ambulance housing stations
- Hospitals and medical clinics
- Convalescent homes

Lifeline and emergency service structures are facilities which must remain operable in the event of a major geologic disruption as well as during everyday activities. Special design and construction considerations are, threfore, warranted to assure that these facilities remain operational during and after all types of natural and man created disasters. Because of the nature of certain lifeline and critical facilities, additional precaution should be taken to protect life, property, and certain essential activities from the disruption of the facility itself. Disruption of gas or liquid fuel lines, contamination of potable water supplies because of broken sewage lines, or release of large bodies of water contained by a dam are just a few examples of such problems. Duke (1975) summarizes several serious implications associated with failure of lifeline and emergency service structures, considered from the viewpoint of their impact on society.

- Direct financial losses
- The inability to protect against secondary disasters such as fires, famines, epidemics, and crime.
- The inability to protect against the lifeline itself if its nature is such that it may become a hazard to life and property.
- Suspension of employment and other activities which are dependent upon the lifeline.

It is important that high occupancy and certain other important public facilities be designed to provide public safety. These structures must be analyzed, designed, and constructed to minimize the level of risk associated with geologic hazards. Although these structures may not be required to remain operable after a disruption, a serious failure of the structure clearly poses a threat to loss of life, property, and human activity. Also, during this study, the following types of existing and proposed high occupancy and important public facilities were reviewed from a geotechnical viewpoint:

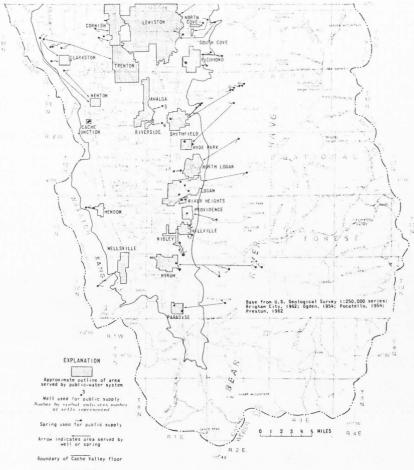
- Schools
- Large apartment complexes
- Office buildings higher than 3 stories
- Dams higher than 20 feet (6 meters) or retaining more than 50 acre-feet (61,700 cubic meters) of water.

Many types of geologic processes act slowly to degrade engineering works. Proper planning, analysis, design, construction, and appropriate consideration of the natural forces at work will greatly minimize the problem.

Characteristics of Cache County

Cache Valley, Utah, is a geographical subdivision lying within the bounds of Cache County and is located in the northeast corner of Utah. Figure 1-1 shows the county boundary and indicates the approximate limits of Cache Valley, Utah. Figure 3-14 shows the location of Cache County within the bounds of Utah. Box Elder County, a much larger county lies to the west, and Rich County, a smaller county, lies to the east.

Cache County is one of 29 Utah counties. It ranks 5th in population and 23rd in size among the other counties. The 1970 census indicates that about 42,300 people reside in Cache County and Logan is the largest city with a population of approximately 24,000. Most of the people live in towns and rural communities but many live on farms.



Boundary of Cache Valley drainage basin

Figure 1-1. Location of Cache County communities and the sources of public water (after Bjorklund and McGreevy, 1971).

Agriculture and light industry provides the primary economic activity of the county. Most of the valley land, up to the foothills, is irrigated and farmed and in many areas where topography and slope permits, the foothill benches are dry-farmed.

Utah State University is located at Logan and was founded in 1888. The university is widely recognized as one of the leading institutions in Agriculture and Water Resource Technology. Its presence plays an important part in the economic and cultural life style of the county.

Cache County's climate is moderate and dry. Average summer maximum temperatures range from 80°F to 90°F and average winter conditions are cold. The valley normally receives approximately 20 inches (50.8 cm) of precipitation annually with the accumulations of up to 50 inches (127 cm) normally occurring in the mountains (Jeppson et al. 1968). Snow usually covers the valley floor during December, January and February. The frost-free season is usually about 150 days, lasting from May through September (Bjorklund and McGreevy, 1971).

Purpose of Study

For this study, geologic hazards were divided into two groups; seismic hazards and other geologic hazards. These hazards, some of which may influence many local vital facilities, are listed below and are described in detail in Chapter 3.

Seismic hazards

- Strong ground shaking
- Surface fault rupture
- Liquefaction

- Miscellaneous ground failure
- Seiches

Other geologic hazards

- Landslides
- Flooding
- Erosion
- Expansive soil
- Subsidence

The purpose of this investigation was to study the level of geotechnical and geologic expertise that was involved in the siting, design, and construction of existing critical public facilities, and to study geologic hazards threatening existing structures in Cache Valley, Utah, in an effort to provide insight to improve future construction. This study was intended to be indicative of the adequacy of current design practices in Cache County.

Geologists and geotechnical engineers should both play an important role in the siting, design, and construction of vital public facilities. The geologist is involved in the study of geology which Legget (1962) defines as:

> "That branch of natural science devoted to the study of the physical features of the earth, the composition and structure of the rocks composing it, the forces at work in altering it, and the record of the animals and plants that have lived on its lands and inhabited its seas." (Legget, 1962)

The geologist's input into the planning of engineering works is based largely on observation. Geologic events that have transpired in the past are studied in an effort to define the physical nature, composition, and forces which may act on earth materials in the future. The role of the Geotechnical Engineer is to study the earth materials and their ability to support engineering works. The Geotechnical Engineer often uses tests on the soil and various analytical procedures to help predict future behavior. The chief difference between the fields of geology and geotechnical engineering is that in geology, observation predominates and in geotechnical engineering, testing and analysis predominates. Competency in these disciplines requires experience. Proper evaluation of a site should incorporate a planning, testing, and analysis program which is based on a team effort involving both a geologist and a geotechnical engineer.

Methodology

Most of the facilities reviewed during this study are owned and operated by communities within Cache County, Utah. Table 1-2 is a list of communities involved and the approximate population of each (1970 census). Figure 1-1 by Bjorklund and McGreevy (1971) shows the location of the communities and indicates the sources of public water.

Basic data was collected on existing and proposed vital facilities. This data was studied to assess the level of geotechnical input and to identify problem areas associated with geologic hazards. Collection of the basic data for each facility consisted of a detailed site examination of the facility and its environment, conversations with persons known to be knowledgeable about the design or construction of the facility, reviews of published environmental and geotechnical reports pertaining to the facility and reviews of all available plans,

specifications, design or construction notes, and reports on the facility. In some cases, especially those involving older structures, little or no information or documentation was available to indicate design considerations or thoroughness of the geotechnical investigation. Accurate and meaningful collection of facility data was largely dependent upon the cooperation received from the design professionals, owners, and persons knowledgeable about the facility. If their cooperation was lacking or if the information was not available, the evaluations of the facility may have reflected less geotechnical design and construction considerations than actually may have been addressed.

Community	Population
Amalga	220
Benson	90
Clarkston	550
Cornish	200
Cove	50
Hyde Park	1200
Hyrum	2400
Lewiston	1300
Logan	24000
Mendon	365
Millville	450
Newton	470
Nibley	380
North Cove	50
North Logan	1500
Paradise	420
Providence	1700
Richmond	1050
River Heights	1050
Smithfield	3500
Trenton	400
Wellsville	1300

Table 1-2. List of communities included in this study.

The state-of-the-art in geotechnical engineering has advanced significantly in the last several decades. Accordingly, the design standards for most facilities constructed in recent years were much more rigorous than the standards used during the construction of older structures. Since this evaluation involved the study of geotechnical considerations given to the planning, design, and construction of various facilities in light of current standards, present day standards provided the basis for comparison. It was also recognized that certain types of structures require special considerations and individual attention. The design of a vital facility such as a hospital warrants a very detailed and thorough geologic and geotechnical investigation. In contrast, a culinary water storage facility would not warrant the detailed site investigation that a hospital would require. However, these are both critically important facilities which must be designed to assure a high level of dependability.

In reviewing individual facilities, a subjective judgment was required to evaluate the level of geologic and geotechnical expertise which should have been involved in the siting, design, and construction of the facility. Guidelines, as set forth by various state and Federal agencies, private design professionals, and universities, were used in making these judgments.

Chapter 3 discusses many of the important geologic hazards that could affect individual facilities and some of the guidelines that should be employed during siting, design, or construction of a facility. In carrying out the study, the actual geologic and geotechnical expertise that was employed with the design and construction of a

facility was compared with the subjective judgment of what should have been required for the facility. A discussion of apparent hazards affecting some facilities is included in Chapter 4. The purpose of the discussion of Chapter 4 is to indicate that certain geologic hazards do exist in Cache County and may influence the safety of some critically important structures. The discussion of Chapter 4 is not intended to reflect negatively on any community or organization nor is it intended to indicate that better alternatives to siting design or construction were available in all cases. It is intended to indicate that certain geologic hazards may pose a threat to vital utilities and structures and that the hazards must be acknowledged and studied as necessary for the safe operation of the facility.

This study considered only the geologic and geotechnical features affecting a site and did not evaluate such factors as hydraulic design, political or social impacts, or economic considerations. The proper siting, design and construction of a facility must consider all of these factors.

CHAPTER 2

GEOLOGIC SETTING

General Features

Cache Valley is a north-south trending elongate basin located partly in Utah and partly in Idaho. The valley floor is about 60 miles (97 kilometers) in length with approximately 35 miles (57 kilometers) in Utah and 25 miles (40 kilometers) in Idaho. The width varies roughly from 8 to 16 miles (12.8 to 25.7 kilometers), with the widest point being at the Utah-Idaho border. The valley's drainage basin which is part of the Bear River Basin, includes approximately 1,840 square miles (4765 square kilometers). The lowest point in the valley, where the Bear River exits at Junction Hills, is about elevation 4,400 feet. Mountain ranges surround the county on three sides; the Bear River Range to the east, the Bannock, Malad, and Wasatch Ranges to the west and the South Hills to the south. The valley floor is at an approximate elevation of 4500 feet above sea level and is principally composed of lacustrine sediments deposited in the ancient Lake Bonneville which once flooded the valley.

Cache Valley is a complex graben, bounded by basin and range faults on both sides. The valley is the downthrown fault block and is composed of sediments of Cenozoic Age. The uplifted blocks forming the surrounding mountain ranges are composed mainly of rocks of Paleozoic Age and limestone and sandstone are the predominant rock types. The faults on either side of the valley are steeply dipping normal or gravity faults and are recognized as active. (Refer to Figure 2-1 by Williams, 1962.) The vertical displacements along these fault systems may exceed 10,000 feet (3048 meters) in some locations (Bjorklund and McGreevy, 1971). This horst and graben geologic structure is typical of the basin and range Physiographic Province.

Utah is seismically active and is part of the Intermountain Seismic Belt. Principal active fault zones such as the Cache Valley Faults and the Wasatch Fault are part of this sesimic belt and are discussed in detail in Chapter 3.

Lake Bonneville

During and subsequent to the fault movements the valley has continually received clastic sediments which have backfilled the valley to its present level. The deposition of these sediments was largely controlled by the ancient Lake Bonneville which once occupied the valley. Lake Bonneville existed during the Pleistocene epoch and covered a major portion of Utah and extended into Idaho and Nevada (see Williams, 1962, pp. 131). The Great Salt Lake is the present remnant of the ancient lake and is now approximately 950 feet (290 meters) lower than the maximum Bonneville elevation. The elevation of the present Great Salt Lake has not risen higher than about 110 feet (34 meters) above the current level for over 11,000 years (Williams, 1958).

The level of Lake Bonneville is known to have fluctuated over a wide range of elevations. Two distinguishable levels can be traced in the Cache Valley. The lake is thought to have risen to an approximate elevation of 5135 feet during the maximum cycle, which is known



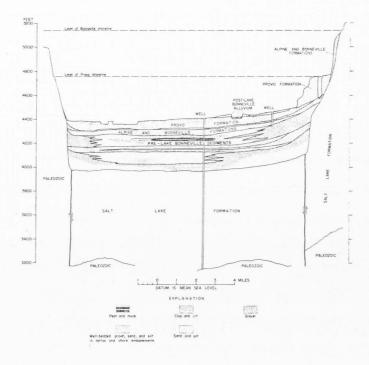


Figure 2-1. Diagrammatic cross-section of Cache Valley from Cache Butte through Amalga to Smithfield (after Williams, 1962).

as the Bonneville level. The lake then receded to a lower level and possibly even dried up for a short geologic time before returning to the Provo level. The Provo level is found at about elevation 4800 feet. Due to fault displacements subsequent to the development of these two beach terraces, these levels are not consistent throughout the valley. It has been indicated that the Bonneville level differs as much as 100 feet (30 meters) at different locations within Cache Valley (Bjorklund and McGreevy, 1971). It is also suggested that the difference in shoreline elevations may be partly due to isostatic rebound due to unloading as Lake Bonneville receded (Bjorklund and McGreevy, 1971).

The Bonneville level sediments cover many of the higher bench areas up to the foothill areas. These sediments range in thickness from about 50 to 100 feet (15 to 30 meters) and are the oldest of the Lake Bonneville unconsolidated sediments. They are composed mostly of silt but also include some gravel in embankments and small deltas.

The Provo level sediments are found extensively in the Cache Valley and they cover the older Bonneville level sediments in many areas. The Provo formations include extensive gravel deposits in spits and river deltas, as well as lacustrine silt and clay sediments that settled from suspension in the lake water onto the lake bottom. The thickness of this formation is as much as 75 feet (23 meters) in some locations but varies from 50 to 75 feet (15 to 23 meters) in thickness (Williams, 1962).

Provo level deposits are most important in the Cache Valley, since they make up the sand and gravel deposits found in this area.

The east side of the valley illustrates this fact, since each of the principle streams entering the valley has formed large sand and gravel delta deposits at the mouths. This can be seen at the mouths of the Little Bear River, Blacksmith Fork, Providence, Logan, High Creek, and Cherry Creek canyons. These large deltas, partly reworked by the parent streams as Bonneville Lake receded from the Provo level, produce the terraced land forms as seen today. Utah State University stands on the major Provo level bench, and the city of Logan occupies three or more of the small recessional benches.

The towns of Lewiston, Cornish, Trenton and Benson have been built on a large sand levee along the Bear River. This levee was formed as the river followed its course during the time when the water was receding from the Provo level. The sand was derived from a large sand delta at the north end of the valley which was formed when the lake stood at a higher elevation. As the lake receded, the delta was disected and the sediments were carried over the new flood plains, toward Junction Hills, where the river presently exits from the valley.

Groundwater

Groundwater is generally abundant in the valley. The source of the water is mainly from the Paleozoic rocks of the mountains, which feed sand and gravel deposits within the valley. Solution within Paleozoic carbonate rocks of the Bear River Range has developed zones of high permeability, especially along joint systems and bedding planes. Although a few large caves have developed as solution channels in these rocks, large solution channels are not found abundantly.

Within the valley areas, below the foothills, most of the developed groundwater systems derive their water from wells. However, where coarse alluvium lies above Lake Bonneville silt and clay deposits contact springs are common. For instance, the major river deltas of the Provo sediments such as the Logan River and the combined deltas of Blacksmith Fork and the Little Bear River lie above the fine lacustrine lake sediments. The groundwater moving into the valley from the mountain sources travels in the pervious zones, above the fine grained lacustrine deposits. Similarly, where post Lake Bonneville deposits extend out on the old lake bed the springs are found to occur along the sides and ends of the deposit.

Typical of lacustrine deposits and the fluctuating level of Lake Bonneville, inter-tongued layers of silt and clay, and sand and gravel was deposited along the sides of the valley in the subsurface strata. The coarse grained sediments in these layers store large quantities of groundwater and provide a high enough permeability that the water is readily available if tapped. Many of these large underground aquifers are confined from near the recharge areas of the foothills to far out into the valley and consequently, widespread artesian conditions are prevalent. These aquifers are probably sheets and channels of coarse grained sediments which grade valleyward into sand (refer to Figure 2-2 by Bjorklund and McGreevy, 1971). The illustration in Figure 2-2 shows the confining nature of the fine-grained sediments and provides an idealistic portrayal of the movement of groundwater from the mountains to the valley. Piezometric heads of as high as 62 feet (18.9 meters) above the land surface have been reported near

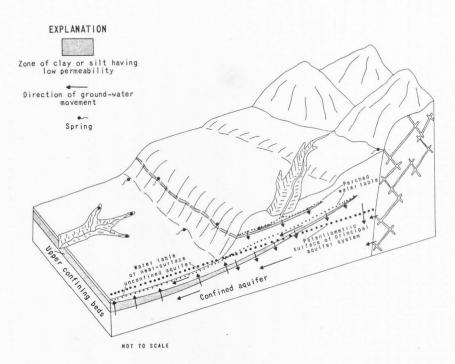


Figure 2-2. Relation of confined, unconfined, and perched groundwater in Cache Valley. (After Bjorklund and McGreevy, 1971.)

Benson (Bjorklund and McGreevy, 1971) however, the head on most wells is 40 feet (12 meters) or less. The confined aquifers of the valley commonly show the deeper beds having slightly higher piezometric heads. This may be due to differences in the permeability of various aquifers and the associated head loss. Also, the recharge area for most aquifers is complex in nature and elevation differences in the source water may influence the varying piezometric level of the different aquifers. The bulk of the artesian conditions generally occurs below the valley floor elevation of 4500 feet (Bjorklund and McGreevy, 1971).

Many springs occur along the base of the mountains and are probably fault controlled. These springs are generally located in a linear fashion along the fault traces and may provide for lateral movement of groundwater along the fractured fault zone. Although the groundwater temperature fluctuates a minor amount throughout the valley from these apparent fault controlled springs, the temperature fluctuation is not enough generally to suggest that the source of the water is deep. Instead, the fault zones may act primarily as a zone for lateral conveyance of the groundwater from the higher mountain sources.

Perched water tables exist in many parts of the Cache Valley region. Several perched water tables can conceivably exist at one location and are formed as the water which percolates downward, is intercepted by soils of lower permeability. The water inflow, above the zone of low permeability is greater than the transmissibility of the fine-grained soil. Perched water tables are encountered in excavations for construction and in wells but are poorly mapped due

to seasonal groundwater level fluctuations and lack of wells which tap these zones.

The Bjorklund and McGreevy (1971) report indicates that there has been no upward or downward trend for the groundwater levels in the valley in the 30 years of record, as recorded from wells (other than response to precipitation). Increased amounts of groundwater and overland flow has been diverted for irrigation purposes in recent years. This water is released into perched surface water tables and may be causing a rise in the near surface water table in some parts of the valley.

The Bear River delta is significantly different than many other river delta areas in Cache Valley. The deltas of other streams which enter the valley, are formed by geologically young streams with sufficient energy to transport coarse-grained sediments. The Bear River is an older stream and has a lower energy level. This river delta has deposited sand and silt which is derived from upstream basins, and generally contains poor aquifers.

Post Lake Bonneville

Relatively little geologic change has taken place since Lake Bonneville occupied the valley. The streams entering the valley have continued to cut into their respective deltas and the flood plains and deltas have been reworked and redeposited further out on the old lake bottom.

CHAPTER 3

GENERAL ASPECTS OF GEOLOGIC HAZARDS

Seismic Hazards

General

Earthquakes can cause damage in a variety of ways. This damage can be mitigated only by properly understanding seismic mechanisms and through appropriate planning, zoning, design and construction in hazardous areas. The major seismic hazards considered in this study are described in this chapter and include:

- Strong ground shaking
- Surface fault rupture
- Liquefaction
- Miscellaneous ground failure
- Seiches

Earthquake scales

Intensity and magnitude are terms which are commonly used to describe certain characteristics of an earthquake. Intensity is used to indicate earthquake severity at a specific location. It measures the observable effects of an earthquake as determined through interviews in the quake-stricken area, damage surveys, and earth movement studies.

Magnitude is a term which expresses the total amount of energy release from an earthquake as determined by measuring the amplitudes produced on standard recording instruments. Therefore, it does not consider the effects at any specific location. The equation relating magnitude of an earthquake and the amount of energy which it releases is given by Lew, Leyendecker, and Dikkers (1971) as

$$\log_{10} E = 11.5 + 1.5 M$$

where E is the energy and M is the Richter magnitude. As the magnitude increases, the amount of energy released increases logarithmically. An earthquake of magnitude 8 represents an energy release of about 31.6 times the amount of energy release in a magnitude 7 earthquake, and about 1000 times as much energy as a magnitude 6 earthquake.

The Modified Mercalli Intensity Scale is shown in Table 3-1. The Richter Scale of Magnitude versus equivalent energy in tons of TNT is illustrated in Figure 3-1. The energy released from three local earthquakes are shown on Figure 3-1 and compared with five well known earthquakes and with the energy released from both the Hiroshima Atom Bomb and 1 megaton H-Bomb. Also, shown on Figure 3-1 are estimates of the energy that would be released from maximum credible earthquakes along the Wasatch Fault and the East and West Cache Faults. The maximum credible earthquakes were estimated from data presented by Bonilla (1967).

Strong ground shaking

<u>Characteristics of bedrock motion</u>. Ground shaking during earthquakes causes the most widespread damage as evidence from recorded earthquakes. When a fault zone ruptures, soil and rock materials

Intensity	Description
Ι.	Not felt.
II.	Felt by persons at rest, on upper floors, or favor- ably placed.
III.	Felt indoors. Hanging objects swing. Vibration like passing of light trucks. May not be recog- nized as an earthquake.
IV.	Hanging objects swing. Vibration like passing of heavy trucks or sensation of a jolt like a heavy ball striking the walls. Standing motor cars rock. Windows, dishes, doors rattle. Glasses clink. Crockery clashes. Wooden walls and frames creak.
v.	Felt outdoors; direction estimated. Sleepers waken ed. Liquids disturbed, some spilled. Small unstab objects displaced or upset. Doors swing, close, open. Shutters, pictures move.
VI.	Felt by all. Persons walk unsteadily. Windows, dishes, glassware broken. Knickknacks, books, etc. off shelves. Pictures off walls. Furniture moved or overturned. Weak plaster and masonry D cracked. Small bells ring (church, school). Trees, bushes shaken visibly, or heard to rustle.
VII.	Difficult to stand. Noticed by drivers of automo- biles. Hanging objects quiver. Furnature broken. Damage to masonry D, including cracks. Weak chimneys broken at roof line. Fall of plas- ter, loose bricks, stones, tiles, cornices, also unbraced parapets and architectural ornaments. Some cracks in masonry C. Waves on ponds; water turbid with mud. Small slices and caving in along sand or gravel banks. Large bells ring. Concrete irrigation ditches damaged.

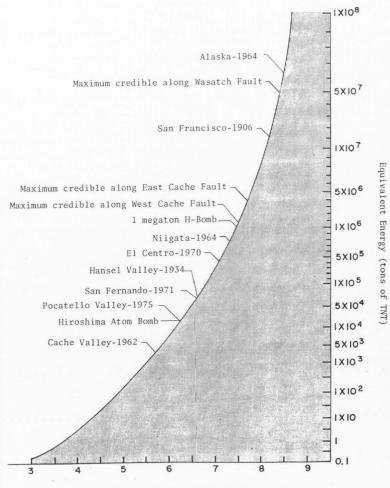
Table 3-1. Modified Mercalli intensity scale of 1930. (Abridged and rewritten.) (After Lew, Leyendecker, and Dikkers, 1971.)

Table 3-1. Continued.

Intensity	Description
VIII.	Steering of automobiles affected. Damage to masonry C; partial collapse. Some damage to masonry B; none to masonry A. Fall of stucco and some masonry walls. Twisting, fall of chimneys, factory stacks, monuments, towers, elevated tanks. Frame houses moved on foundations if not bolted down; loose panel walls thrown out. Decayed piling broken off. Branch es broken from trees. Changes in flow or temperature of springs and wells. Cracks in wet ground and on steep slopes.
IX.	General panic. Masonry D destroyed; masonry C heavily damaged, sometimes with complete collapse; masonry B seriously damaged. General damage to foundations. Frame structures, if not bolted, shift- ed off foundations. Frames racked. Serious damage to reservoirs. Underground pipes broken. Conspicuou cracks in ground. In alluviated area sand and mud ejected, earthquake fountains, sand craters.
х.	Most masonry and frame structures destroyed with their foundations. Some well-built wooden structures and bridges destroyed. Serious damage to dams, dikes embankments. Large landslides. Water thrown on banks of canals, rivers, lakes, etc. Sand and mud shifted horizontally on beaches and flat land. Rails bent slightly.
XI.	Rails bent greatly. Underground pipelines completely out of service.
XII.	Damage nearly total. Large rock masses displaced. Lines of sight and level distorted. Objects thrown into the air.
	Quality of Masonry (Brick or Other)
Masonry A	Good workmanship, mortar, and design; reinforced especially laterally, and bound together by using steel, concrete, etc.; designed to resist lateral forces.
Masonry B	Good workmanship and mortar; reinforced, but not designed in detail to resist lateral forces.

Table 3-1. Continued.

Quality of Masonry (Brick or Other)		
Masonry C	Ordinary workmanship and mortar; no extreme weaknesses like failing to tie at corners, but neither reinforced nor designed against horizontal forces.	
Masonry D	Weak materials, such as adobe; poor mortar, low standards of workmanship; weak horizontally.	



Richter Scale of Magnitude

Figure 3-1. Comparison of Richter magnitude versus equivalent energy of TNT (1 ton = 908 Kg).

transmit the associated vibrations by means of waves propagating through the earth's crust. Four basic types of seismic waves cause ground motion:

- Primary wave (P wave)
- Secondary wave (S wave)
- Love wave (Surface wave)
- Rayleigh wave (Surface wave)

The P wave is a longitudinal or compression wave which travels at about 4 miles per second (6.4 kilometers per second) (Lew, Leyendicker, Dikkers, 1971). Particle motion for a P wave is in the direction of the wave propagation. S waves travel about half the velocity of P waves. Particle movement of an S wave occurs at right angles to the direction of propagation. The Love and Rayleigh wave travel about 2.5 and 2.25 miles per second (4.0 and 3.6 kilometers per second) respectively. These waves travel only along the earth's surface and have much longer periods than S and P waves. Love waves produce lateral shear forces in a horizontal plane and Rayleigh waves produce an elliptical motion similar to wind driven ocean waves (Lew, Leyendicker, Dikkers, 1971).

The P and S waves pass through dense subsurface layers of the earth's crust (bedrock) and then propagate vertically upward from the bedrock through less dense soil deposits to cause horizontal ground shaking at the ground surface. Figure 3-2 provides a diagramatic sketch of the propagation of seismic waves from the bedrock surface up through the soil profile. Earthquake damage is caused by the horizontal motion at the base of structures. This motion can have

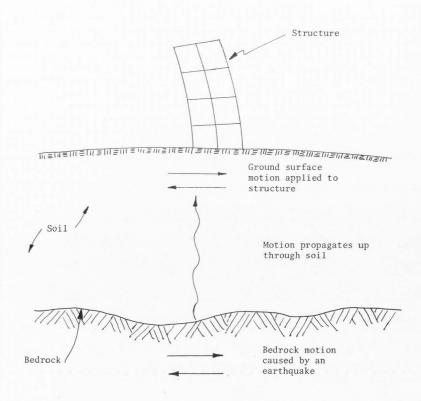


Figure 3-2. Diagrammatic sketch of the propagation of motion through soils during ground shaking.

a detrimental effect on structures located many miles from the causitive fault.

<u>Characteristics of ground surface motion</u>. The characteristics of the ground surface motion to be expected at any particular site is dependent on several factors:

- The amount of energy released during the earthquake which may be described by the Richter Magnitude.
- The distance from the site under consideration to the zone of rupture of the earthquake.
- The character of the site subsurface conditions.

The amount of energy released during a seismic event has been related to the length of the fault rupture (Bonilla, 1967). This provides an estimate of the maximum credible earthquake along a fault by assuming that rupture could occur along the entire fault. Smaller magnitude earthquakes may occur if rupture only occurs along a portion of the fault. Also, the amount of energy released may be very dependent upon the state of stress along the shear zone and the elastic properties of the rock. Rock types which allow the accumulation of strain energy under increasing shear stress, can release violent vibrations when movement in the fault zone eventually occurs.

The distance between the site and the causitive fault and the magnitude of an earthquake influence the characteristics of the bedrock motion. Both the maximum acceleration and the predominate period of the bedrock motion are a function of the distance from the causitive fault. For an earthquake of any given magnitude, maximum accelerations will generally decrease with increasing distance from the causitive fault zone while predominant periods of motion will increase (Kiefer, 1968; and Seed, Idriss, Kiefer, 1969). The maximum bedrock acceleration and the predominate period of the bedrock motion both increase with earthquake magnitude.

While the above relationships generally apply for bedrock motion the response at a specific site is dependent on the properties of the subsurface profile. Kiefer (1968); Seed and Idriss (1969); and Seed, Idriss and Kiefer (1969) discuss the characteristics of ground motion during earthquakes in more detail. Seed and Schnabel (1972) discuss accelerations in rock and the various effects for areas of the Western United States.

Damage potential from ground shaking. Local soil deposits may greatly modify the intensity of the ground motion, as measured by the maximum acceleration and the fundamental period of the motion. Experience has been gained by studying high shaking damage in areas of San Francisco, Santa Rose, and San Jose, during the 1906 San Francisco Earthquake and parts of Caracas, Venezuela, during a 1967 earthquake. The following accout from Seed (1975b) illustrates how soil conditions may effect the ground motion characteristics during earthquakes.

During the Caracas Earthquake, which had a Richter Magnitude of about 6.4 and a fault zone located about 35 miles (56 kilometers) from the city, intense shaking caused the total collapse of four 10to 12 story apartment buildings. A correlation of depth of underlying soils to structural damage showed that for 3-to-5 story buildings located on soil depths of 98 to 164 feet (30 to 50 meters), the damage was many times greater than for similar structures located where soil depth was 328 feet (100 meters) or more. For 5-to-9 story structures, the most damage was found where the soil depths ranged from 164 to 230 feet (50 to 70 meters). However, for structures greater than 10 stories, the structural damage was several hundred percent greater where the soil depths underlying the structure exceeded 525 feet (160 meters) than for soil depths less than 459 feet (140 meters). This study demonstrated that soil conditions have a significant affect on the ground motion characteristics, even for the same earthquake and within the same city (Seed, 1975b).

The response of a structure to ground shaking is largely dependent upon the relationship between the fundamental period of the building and the predominate period of the ground surface motion. The predominate period of the ground surface motion for a given earthquake is a function of the fundamental period of the site. A critical state can develop when the fundamental period of the site. A critical state can develop when the fundamental building period matches the fundamental period of the site. Both theory and experience indicates that the fundamental site period is dependent on the intensity of the bedrock motion, sediment thickness, soil firmness, and degree of saturation. The fundamental period of the building is related to its weight, material properties, geometry, and structural details. Taller buildings generally possess longer fundamental periods as to deep, saturated, soil deposits. Short rigid structures generally exhibit short fundamental periods as do shallow, very dense, or rocky subsurface conditions. This suggests that the damage potential of an earthquake would be

maximized when tall flexible buildings are located on deep soft soil deposits or low rigid buildings are located on shallow stiff soil deposits. Thus, the damaging affects of an earthquake are maximized when there is a similarity in the natural periods of the structure and the ground which it rests on. The 1976 Uniform Building Code considers the relationship between the fundamental site period and the fundamental period of the building in establishing the design forces to be used for the design of the building. The ratio of these two periods is used as a lateral force coefficient in establishing the lateral design force which acts on a structure.

Figures 3-3 through 3-7 from Lew, Leyendecker and Dikkers (1971) are examples of the type of destruction that can result from earthquake ground shaking hazards. These photographs were taken of the aftermath of the 1971 San Fernando Valley Earthquake. Earthquakes of an equal or greater magnitude than the 1971 San Fernando Valley earthquake can be expected in Utah, and the earthquake resistance of structures in Utah should not be expected to be any greater than those of Southern California. Indeed, because of the extensive amount of masonry construction, the resistance of many structures in Utah may be considerably less than those of Southern California. Thus, earthquake damage in Utah due to ground shaking is a very serious geologic hazard. The type of destruction in Utah from a moderately strong earthquake such as occurred in the San Fernando Valley in 1971, can be expected to be very similar to that as shown in Figures 3-3 through 3-7. The fact that earthquakes of a greater magnitude than the San Fernando Earthquake could likely occur in Utah and the fact that such a high



Figure 3-3. Olive View Medical Center, showing extensive damage resulting from the 1971 San Fernando Earthquake (after Lew, Leyendecker, and Dikkers, 1971).

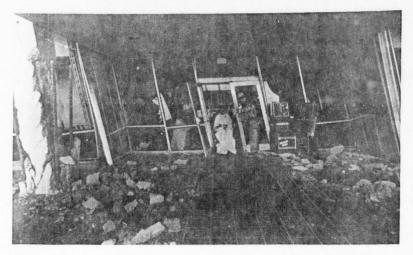


Figure 3-4. Lateral displacement of first floor of Olive View Medical Center (after Lew, Leyendecker, and Dikkers, 1971).



Figure 3-5. Main business district of the city of San Fernando (after Lew, Leyendecker, and Dikkers, 1971).



Figure 3-6. Total collapse of brick masonry wall surrounding three-story wood frame building (after Lew, Leyendecker and Dikkers, 1971).

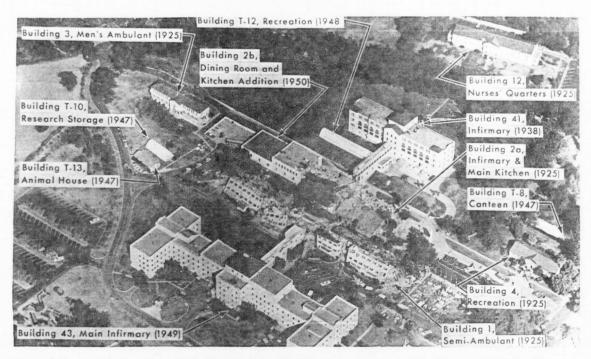


Figure 3-7. Aerial view of the San Fernando Veterans Administration Hospital. Collapsed building (No. 1 and 2a) constructed in 1925, where 46 patients and hospital workers were killed. Many of the buildings, also constructed in 1925 (buildings 3, 4, and 12), although still standing, were severely damaged (after Lew, Leyendecker and Dikkers, 1971). percentage of the developments in Utah are in very close proximity to major active faults makes the earthquake hazard from ground shaking in Utah very high.

Surface fault rupture

A second well known seismic hazard is that of fault displacement and surface rupture. If a structure is located astride a fault where movement occurs, it is highly probable that total destruction will result. The design of such a structure, to withstand the destructive forces that are induced by displacement or rupture, would be very complex and costly. Thus, especially for the siting of lifelines, emergency service structures, high occupancy structures and certain other important public facilities, it is imperative that both a professional geologist and a professional geotechnical engineer review the site in detail. This should help assure that the facility will not be placed at a location where faulting is probable.

Many faults are known to be inactive and should not be of as much concern to development as active faults. The definition of an active fault, however, is subject to varied opinion. For example, the Nuclear Regulatory commission places a higher risk on the possibility of renewed movement along an old fault for the design of a nuclear facility than would be considered for the design of a relatively minor structure. The type of land use and the importance of the structure should govern the consideration given to possible movement along a fault. An active fault can be defined as one that has moved in recent geologic times. Many geologists would include at least the Holocene Epoch in this period, or about 10,000 years (±) (Wiegel, 1970). Movement may occur along one fracture surface or along a wide zone involving many fractures (fault zone). Each fracture shear zone may range in width from less than an inch to many tens of feet. The fracture itself is commonly not a plannar surface but rather a curved and irregular surface of sliding. A fault trace may commonly exhibit an en echelon¹ pattern of rupture, however, this is dependent upon the state of stress acting in the crust of the earth and the type of earth materials.

Movement along fracture surfaces can be in virtually any orientation. In order to distinguish the type of movement of faults, a steeply dipping fault is classed as a high angle fault, while a shallow dipping fault is known as a low angle fault. If movement of two points is roughly along strike, it is classed as a strike-slip fault and if movement is along the dip, it is classed as a dip-slip fault. Faults of the Basin and Range Province, such as the Wasatch and East Cache Faults of Northern Utah, are the dip-slip type. The type of movement along a fault is extremely important in estimating the amount and type of damage that might be expected to occur.

As discussed previously, movement commonly does not occur along a single fault plane, but rather within a fault zone. Any one of many fracture surfaces may slip. Generally, within such a fault zone, an area of "most likely" slippage may predominate. Often, short secondary faults may branch out of the main fault zone. Slippage along these minor faults may only be a fractional of the amount of the slippage

¹en echelon is a term used to describe a group of parallel ruptures, oriented in a steplike pattern, at an angle to the general direction of fault rupture.

along the main fault. Such secondary faults may form an en echelon pattern of rupture from the main fault zone and would usually be no more than a few miles long.

Although faults with many thousands of feet of past total displacement are not uncommon, the displacement probably represents the summation of a very large number of separate movements over a very long period of time. The individual movements are probably no more than a few inches to a few feet. One of the largest single fault displacements known to have occurred in recorded history of the United States, took place near Salt Lake City, Utah. A single vertical fault movement of 20 feet (6 meters) was estimated to have occurred along the Wasatch Fault near the mouth of Little Cottonwood Canyon. This was estimated to have occurred within the past 300 years (U.S. Senate, 1975, pp. 32). Measurements by Bruce Kaliser of the Utah Geological and Mineral Survey indicate that displacements near the area of Little Cottonwood Canyon range from 20 to 45 feet in height, and that the total vertical displacement, as seen in 6 separate adjacent fault scarps, totals about 188 feet (57 meters) (U.S. Senate, 1975, pp. 32). Another large displacement occurred in 1899, at Yakutat Bay, Alaska, where a single displacement amounted to 47 feet (14 meters) (Tarr and Martin, 1912).

Not all faults extend to the surface of the earth. Movements may occur at depth but due to plastic or elastic deformation of the subsurface strata, the displacements may not propagate to the surface. ~ Northern Utah faults have commonly exhibited this type of diastrophism in the recorded history and only a single seismic event in the recorded history of the state has produced surface ruptures. In recent geologic

times, however, many feet of displacement as well as other geomorphic evidence serves to illustrate the geologically recent surface disruptions along all of the major fault zones. Several thousand feet of vertical displacement has occurred during a long fault history in northern Utah.

¹Lifeline facilities, such as gas and water lines, are particularly exposed to fault zone hazards since they frequently must cross active fault zones to service a region. Proper consideration given to the methods of crossing this zone and to the sense of earth movement along the fault zone is helpful in reducing the geologic hazard (Ford, 1977).

Geologic investigations may include a regional analysis to determine the type of movement along a fault, the probable fault patterns and locations, and the relative degree of activity along a fault. Seismic or micro-seismic earthquake records will also indicate the degree of activity of a fault zone.

Liquefaction

<u>Cause of liquefaction</u>. When a loose, saturated, fine sandy deposit is subjected to vibratory motion, the sand can liquefy and thus, loose essentially all of its shear strength. Relatively large land masses with ground surface slopes as slight as two or three percent have liquefied and flowed horizontally for great distances. This phenomenon has received much attention recently due to several rather spectacular failures such as the building foundation failures in Niigata, Japan (Seed and Idriss, 1967, 1968); the Turnagain Heights landslide in Anchorage, Alaska, the landslide in Valdez, Alaska (Seed, 1974);

the Juvenile Hall landslide in southern California (Nichols and Buchanan, 1974); and the failure of the lower San Fernando Dam (Seed et al. 1975). Figures 3-8 and 3-9 are photographs take from Seed (1975a) of the Niigata building foundation failures and the Turnagain Heights landslide, Anchorage, Alaska, that resulted from liquefaction. Figure 3-10 from Lew, Leyendecker and Dikkers (1971) shows the liquefaction damage resulting from the partial collapse and near catastrophic failure of the lower San Fernando Dam in 1971.

Sand particles when in a loose dry condition will decrease in volume when disturbed. However, if the voids within the sand mass are filled with pore water, the volume cannot immediately decrease upon disturbance. Instead, the load is transferred to the pore water and there is an abrupt increase in the pore water pressures and decrease in the intergranular stress. During an earthquake the sand is repeatedly disturbed. Each disturbance tends to cause a decrease in void volume as described above. Since a decrease in volume cannot occur immediately, there is a transfer of load from the sand grains to the pore water. The pressure induced in the pore water in this manner is an excess hydrostatic pressure. If the disturbance persists, the pore pressure continues to increase until the intergranular stress in the sand is reduced to zero. This is termed as a condition of "initial liquefaction" (Seed, 1976).

<u>Factors affecting liquefaction potential</u>. Several significant factors affect the development of a liquefaction condition, including soil properties, initial stresses, and the characteristics of the earthquake. The following factors are listed as most important:



Figure 3-8. Foundation failures in Niigata, Japan (1964) (after Seed, 1975a).



Figure 3-9. Turnagain Heights landslide, Anchorage, Alaska (1964) (after Seed, 1975a).

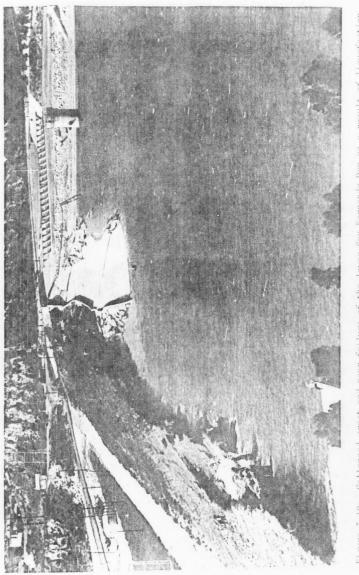


Figure 5-10. Collapsed crest and near failure of the lower San Fernando Dam as a result of liquefaction (after Lew, Leyendecker, and Dikkers, 1971).

- Soil type
- Relative density
- Initial confining pressure
- Intensity of ground shaking
- Duration of ground shaking

Soil type has a significant influence on liquefaction potential. Fine to medium grained sand with a rather uniform grain size has been found to be most susceptible. If the sand contains too many fine particles (minus No. 200 sieve size) or if it is too coarse, the liquefaction potential may not be great.

Relative density of the sand is very significant. Very loose sand is highly susceptible to liquefaction while dense sand has a very low liquefaction potential.

Investigations have shown that the stress required to initiate a liquefaction condition increases with the initial confining pressure. This effect has been shown in field cases such as at Niigata where soil under nine feet of embankment fill remained stable, while similar soils surrounding the embankment liquefied (Seed, Lee, and Idriss, 1967).

The intensity of the ground motion caused by an earthquake is also a very important factor. The susceptability of soil to liquefaction under a given confining pressure and relative density is related to the magnitude of shear forces induced by the earthquake. This effect can again be illustrated by the Niigata earthquake which had a Richter magnitude of 7.3. Records of the past 370 years indicate only slight liquefaction problems at Niigata. During this period, 22 of the major earthquakes affecting the city produced calculated values of ground surface acceleration of 0.12 g or less. Of these, only two earthquakes caused observable liquefaction near Niigata. These two earthquakes are believed to have caused ground surface accelerations of about 0.13 g.

It wasn't until 1964, when maximum recorded accelerations of 0.16 g occurred, that extensive liquefaction in Niigata was observed. Thus, even though a long history of seismic stresses had occurred at the site, they did not cause a critical condition until the 1964 earthquake (Seed, Lee, Idriss, 1967).

Duration of strong shaking and consequently, the number of significant stress cycles to which the soil is subjected influences the extent of the build-up of excess hydrostatic pore pressures. Even though this process begins at the onset of the ground motion, instability will not result until a state of "initial liquefaction" is reached or the intergranular strength becomes less than the forces acting on the land mass. The Turnagain Heights landslide in Alaska did not occur until after approximately 90 seconds earthquake shaking. At this time the stability of the mass had decreased sufficiently to allow movements (Seed, 1974).

Another condition affecting the liquefaction potential of sand is that of the influence of the seismic history on the sand deposit. Experiments by Seed, Mori, and Chan (1975) showed that disturbed sand specimens with a given density, subjected to cyclic stresses, liquefied much sooner than similar undisturbed sand subjected to repeated minor seismic events prior to failure. In a test intended to produce

a simulated seismic history, cyclic stresses were mechanically applied to a remolded sand and water sample. The sample was allowed to build small residual pore pressures, then application of the cyclic stress was ceased. The pore pressure in the sample was then allowed to dissipate completely. This series of events was intended to simulate one seismic event such as an earthquake in which a natural soil deposit is subjected to cyclic stresses but liquefaction does not occur. The procedure described above was repeated five or six times, never allowing the sample to liquefy and allowing complete dissipation of the pore pressure between applications of cyclic stress. Next, the sample was allowed to liquefy and the number of applied stress cycles to failure was recorded. The relative density of the sand during this test changed very slightly from 54.0% to 54.7%. Similar tests were run on sand samples with no simulated seismic history. In these tests, cyclic stresses were mechanically applied to remolded sand and water samples. The pore pressure was allowed to build during the first cycle until liquefaction occurred. The samples with the simulated seismic history withstood eight times as many cycles of shear stress to cause failure as similar sand samples with no seismic history (Seed, Mori, and Chan, 1975).

The effect of seismic history on liquefaction potential is believed to be most pronounced in sands possessing very low relative densities. Therefore, a saturated sand fill having no history of seismic disturbance and having a low relative density, may pose an especially acute liquefaction hazard.

The development of a liquefied zone normally occurs between 10 and 80 feet (3 to 24 meters) below the ground surface. If a liquefied

layer occurs near the ground surface, structures may loose foundation support and tilt or gradually sink downwards as occurred during the Niigata earthquake of 1964. Also, buried tanks may float out of the ground due to buoyant forces (Seed, Lee, and Idriss, 1967).

Frequently, sand boils, sand ridges, and volcanoes develop in a liquefied area after an earthquake. This is caused by the upward movement of ground water from the liquefied layers, carrying sand particles with it. Often the sand will be removed unevenly from the subsurface and the resulting differential settlement of the ground surface can be catistrophic to a building.

Methods to evaluate liquefaction potential. Two methods of evaluating liquefaction potential are commonly used in design practice. One of these methods involves the use of a chart, developed by Seed, Mori, and Chan (1975) which is based on data from performance observations of many sites during past earthquakes. This chart allows the comparison of the ratio of shear stress and effective overburden pressure to the standard penetration resistance for any site under consideration. By comparing typical values for the site under consideration to Seed's chart, a degree of liquefaction hazard can be identified. Figure 3-11 is a reproduction of that chart.

The second method requires an evaluation of the level of stress which will be induced in the soil by an earthquake. This stress level must be compared with the level of stress in the soil, necessary to cause liquefaction. The zone where these two curves overlap, as shown in Figure 3-12 is the zone where liquefaction may occur (Seed and Idriss, 1975). Liquefaction; stress ratio based on estimated acceleration
 Liquefaction; stress ratio based on good acceleration data
 No liquefaction; stress ratio based on estimated acceleration
 No liquefaction; stress ratio based on good acceleration data

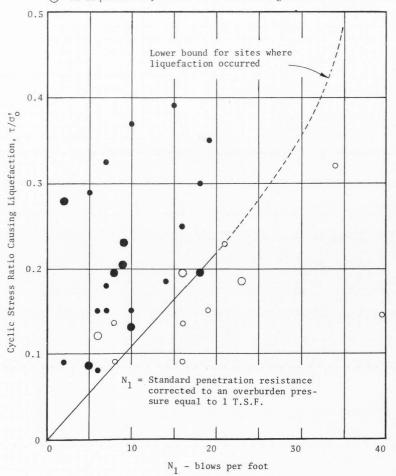


Figure 3-11. Correlation between stress ratio causing liquefaction in the field and penetration resistance of sand (after Seed, Mori, and Chan, 1975).

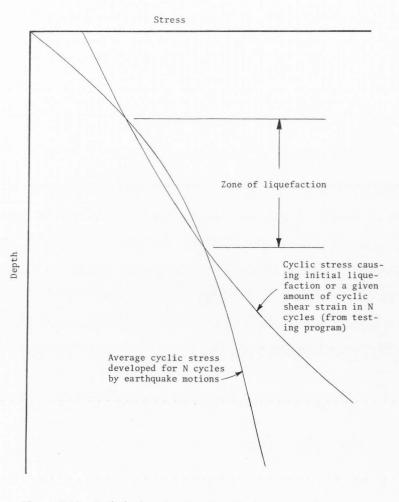


Figure 3-12. Method of evaluating liquefaction potential (after Seed and Idriss, 1971).

Miscellaneous ground failure

Stresses induced in the earth's crust during an earthquake are generally greater than the static forces normally acting on a land mass. Thus, many types of ground failures occur during earthquakes when static forces are augmented by dynamic forces, causing rapid changes in the state of the earth materials. Landslides, ground lurching and cracking, differential settlement, tilting, and loss of strength of loose sand and sensitive cohesive soils are a few of these failures.

Great devastating landslides resulting from earthquakes, have occurred in the past. They can result from faulting and in affect, the natural oversteepening of slopes, or simply by loss of strength upon application of stresses induced by an earthquake. Liquefiable soils or sensitive cohesive clays are examples of soils that can loose strength during an earthquake. Sensitive clays can loose strength due to particle rearrangement when disturbed, and fine sands can liquefy as described previously. These failures can occur on relatively flat slopes.

Landslides can also develop days after an earthquake occurrence as a result of disruption of the natural groundwater flow regime. If the natural groundwater flow channels are severed or blocked, changes in the seepage conditions will occur, often resulting in decreased stability. A later section on slope failures offers an additional discussion about landslides.

Rockfall and snow or rock avalanches can be triggered by ground vibrations. Developments located below steep slopes may be threatened by this type of geologic hazard. A site may experience large undulating surface waves especially if it is located near the zone of energy release and is supported by soft saturated soil. These waves may cause cracking and the development of compression ridges in concrete or other stiff construction materials. Cracking, related to earthquakes, may also be found in areas where relatively stiff soil or construction materials are found near the ground surface. During a liquefaction condition, for example, as the subsurface spreads laterally, the surface may tend to move in blocks, sometimes resulting in large cracks.

Differential settlement may arise from removal of fine sand by the upward movement of groundwater after an earthquake. Settlement of this type can cause excessive stresses in structures. Subsidence or tilting of the land surface may also result from compaction of the subsurface soil, lateral spreading of soft soils, or sliding of large fault or slide blocks. Utilites may be the most affected by this type of movement since they may likely cross fault zones. Broken utility lines and inverse hydraulic grades are problems that must be considered.

Seiches

A seiche is an earthquake generated standing wave occurring within an enclosed body of water such as a lake or reservoir. These waves are similar to waves produced in a bucket when it is jarred. Often the runup of these waves can be as great as 20 to 30 feet (6 to 9 meters), as observed at Kenai Lake in the March 27, 1964, Alaska Earthquake (Nichols and Buchanan, 1974). Overtopping of dams can cause damage to life and property due to hydraulic forces and innundation.

Landslides cascading into large bodies of water can cause extensive destruction and large loss of life from inundation. The resulting wall of water can be several hundred feet high as observed in Italy in 1963, when a huge landslide fell into Vaiont Reservoir. Nearly 3000 lives were lost (Nichols and Buchanan, 1974). In 1958, in Lituya Bay, Alaska, a similar wall of water 1720 feet (525 meters) high surged out of a coastal bay washing out trees and inundating a very large area. The bay, fortunately, was uninhabited at the time (Nichols and Buchanan, 1974).

General Geologic Hazards

The siting and foundation design of important facilities used by the public should require inspection and approval by a competent geotechnical person. Only a professional geologist and geotechnical engineer is capable of fully assessing the extent of potential hazards which may affect a structure located at a given site. The nature of the structure and its use should govern the extent of the geotechnical investigation required. The knowledge gained from observed failures and problems which may be attributed to insufficient geotechnical investigation and analysis, may be used as a guide to indicate what geologic considerations are necessary.

General

Some major geologic hazards, in addition to seismic hazards, are listed and described below:

- Slope failures
- Flooding
- Erosion
- Expansive soils
- Subsidence

A complete discussion of these hazards, their causes and consequences, and the various effects of modifying conditions is far beyond the scope of discussion in this thesis; however, brief summary of several of the most common hazards is presented.

Slope failures

<u>Slope failure problems</u>. Slope failures cause extensive financial loss in the United States every year. It is estimated that landslides will cause about ten billion dollars worth of damage in California alone during the years 1970 through 2000 (Burer, 1973). Recognizing and avoiding locations where landslides have occurred is of extreme importance in avoiding the hazard in the future. When economically feasible, areas susceptable to slope failures can be stabilized by properly identifying the nature of the instability and making the necessary corrections. The best way to prevent damage is to avoid construction in these areas.

Certain natural and human activities tend to decrease the stability of slopes. Activities which may contribute to instability include oversteepening or overloading slopes, altering the groundwater flow regime to create or worsen a seepage condition, and removal of vegetative cover. Landslides occur when the driving forces become greater than the stabilizing forces. Oversteepening of slopes has the direct effect of increasing the active forces driving the mass. Overloading a slope with excessive fill also increases the driving forces tending to cause failure.

<u>Categories of slope failure</u>. Slope failures can be classed into many different categories according to type of movement, type of earth materials, and shape of the sliding mass. Rockfalls and soilfalls are land mass movements where the mass travels vertically through air for most of the distance.

Slides may occur in many different varieties. If a slide occurs in rock surfaces, it may be termed a slump if the movement is rotational, or Block Glide or Rockslide if the movement is along a plannar surface. Slope failures in rock are usually controlled by joints, bedding planes or weak layers in the rock mass and the rockslide materials generally come to rest in a very disoriented arrangement.

A slide occurring in soil may be termed a block slide if the mass moves in relatively large undisturbed blocks along a plannar surface. An earthflow is similar; however, it usually involves a rotational surface of sliding.

If slide movement is very rapid, it is generally classed as a flow rather than a slide. High water content characterizes debris flows, sand or silt flows, and mud flows; hence, they usually take the form of a very gentle sloping slide surface.

Rockfall avalanches or debris avalanches may occur in dry conditions, and the mass usually attains high velocities. The soil or rock

mass may move up to several miles during this type of movement. The slide at Turtle Mountain near Frank, Alberta in Canada is an example of such a slope failure. Seventy lives were lost when the crest of this mountain broke off and slid to the valley below in 1903 (Legget, 1962).

Identification of landslides by aerial photography. Aerial photography is a significant aid in the detection of landslides. A photo taken near Avon, Utah, (Figure 3-13 and reported by Cluff, Glass, Brogan, 1974), is an illustration of the valuable information such photos can provide. Curved arrows indicate the location of old landslides. A hummocky and wrinkled surface characterizes the ground which has moved and the headwall of the slide or the scarp that remains can be easily identified. Low sun-angle photography, used in this photo, provides an additional aid in recognizing landslide features by accentuating the surface contour of the ground. This method uses shadowing and lighting effects produced by the sun at a low angle to accent unusual landforms. Figure 3-13 also shows prominent linear shadows produced by breaks in slope and probably indicates fault locations.

Aerial photography methods of studying landforms examine changes in photographic tone, changes in ground texture, the pattern of physiographic features such as drainage, and the shape of unusual topographic features. Observations and conclusions stemming from examination of the photographs are interpreted in terms of typical geologic structure of the landforms under review. Photographic investigation can provide



Figure 3-13. Low-sun-angle view of slides near Avon, Utah, (after Cluff, Glass, and Brogan, 1974).

valuable preliminary information about a site; however, it should never be substituted for an actual site investigation.

Flooding

Cache Valley has had a long record of damaging floods, which occur mostly in the agricultural areas. Major floods have occurred six times since the year 1900 (Corps of Engineers, 1973), and studies show that floods larger than those of the past can occur in the future.

Flood studies have been conducted for the Logan River and for the Blacksmith Fork River in Cache County. Construction of vital public facilities should be prohibited below accepted projected flood levels. The type of facility along with its use and occupancy should govern the acceptable level of flood risk to a structure.

If flood studies have not been conducted at a proposed project location and floods may present a problem, either studies should be conducted or an alternate site should be selected.

Erosion

Damage each year due to erosion is very extensive. Losses in California due to erosion have been estimated to be as high as 600 million dollars during the years 1970 to 2000 (Bruer, 1973).

The civil engineer is constantly faced with problems caused by erosion such as stream and river encroachments on valuable property, the erosion of surface soils by wind and rain, siltation behind impoundments, and working with the product of the erosive forces of the past. Neglecting a consideration of the process of erosion on a site may likely result in total failure even though the structure itself is sound. Removal of the vegetative cover during the construction of civil engineering works enhances the erosion process by allowing wind and water to carry soil particles with it. In many states, very elaborate erosion safeguards are required whenever construction takes place. The State Engineer's office of Virginia and Maryland for instance, require the professional design of erosion mitigation measures for all construction projects. Normally, state agencies govern the soil conservation practices.

Expansive soils

These soils are found mainly in areas of nonsaturated residual soil deposits of fine material containing montmorillonite clay minerals. These minerals are sensitive to changes in humidity and as a result of the addition of water they expand. If construction takes place on soils of this type and later, humidity changes occur which increases the available water supply to the soil, large deformations can result. This causes extensive damage especially if the vertical movement occurs differentially. The maximum expansion occurs when the initial stress or confining pressure is small. It is estimated that expansion of these soils may be as high as five percent (Zeevaert, 1973). When construction takes place on soils of this type, it is essential that appropriate design considerations be taken.

Subsidence

Subsidence is the downward settlement of the ground surface due to consolidation or removal of underlying materials. For many years Mexico City has experienced problems due to subsidence which has been

attributed to the lowering of artesian water pressures from pumping. Lowering of the water table causes consolidation of clay deposits because of an increase in the intergranular pressure in the clay.

Consolidation of clay deposits by the addition of surcharge loading usually results in subsidence in areas of soft saturated clays. As the additional weight is applied, the soil becomes more compact. Subsidence may also be caused by removal of soluble materials in the soil structure.

Removal of oil from subsurface has long been known to be a cause of areal subsidence. The best remedy for this situation may be to replace the oil with fluids to maintain equilibrium.

Parts of Cache Valley are underlain with soft saturated soil deposits. Upon loading, very large settlements have been experienced. Subsidence related to extraction of groundwater or surcharge loading is a very important geotechnical consideration.

Seismicity of Utah

Cache Valley is located in a seismically active region, within the intermountain seismic belt. This zone of seismic activity extends in a north-south pattern from Arizona and southern California, through Utah, eastern Idaho, and western Wyoming, and terminates in northwestern Montana. This belt is more than 800 miles (1287 kilometers) long and up to 60 miles (97 kilometers) in width. Seismicity along the intermountain belt has been characterized by Smith (1974) as having:

- en echelon patterned active normal faults
- earthquakes with generally shallow focal depths
- marked areas of low or no seismicity and other areas of earthquake swarms

The seismicity along this zone is associated with the contact and complex movement of the basin and range physiographic province on the west and the North American plate to the east. Movement and stress relief of the two lithospheres creates fracture zones and generates seismic activity.

The seismic history of Utah has been documented since 1850. The most complete study of Utah earthquakes prior to 1950 is a comprehensive report by Williams and Tapper (1953). Their report was based on earthquake data taken from the earliest Utah newspapers and scientific journals, and they assigned modified Mercalli intensities to each event. Since 1950, as better equipment was developed and seismograph stations were installed, much more reliable earthquake data has become available.

Appendices A and B include information on 824 Intermountain region earthquakes. Appendix A is a list of 609 earthquakes which occured in Utah during the period 1850 through June 1965. The source of data for the period from 1850 through 1949 was Cook and Smith (1967) as taken from a report by Williams and Tapper (1953) and is the most complete list available of early Utah earthquakes. None of the 344 earthquakes listed in their study were recorded by seismograph instruments; however, they were sufficiently intense to be felt and modified Mercalli intensities were assigned based on damage reports. For the period from 1950 through June 1962, data was furnished by the Coast and

Geodetic Survey (CGS), U.S. Department of Commerce as reported by Cook and Smith (1967). This period includes 77 earthquakes which were generally recorded by seismograph instruments and includes earthquakes of magnitude 2.0 and greater. From July 1962 through June 1965, 188 earthquake epicenters were recorded with seismograph instruments by the CGS and the University of Utah (Cook and Smith, 1967) and includes magnitudes 1.4 and greater.

Appendix B includes a list of 418 earthquakes with a magnitude of 2.4 and greater which have occurred within a 186 mile (300 kilometer) radius of Logan, Utah. This list includes earthquake information from 1853 through June 1976, with a great deal of duplication of the listing in Appendix A for the period from 1853 through June 1965. Appendix A is included separately because the Williams and Tapper report is probably the most complete list available of early Utah earthquakes.

Geologists have mapped many surface and subsurface faults in Northeastern Utah. The major faults are shown on Figure 3-14. However, there is no doubt that there are many faults that have not yet been discovered. Several geologically active faults exist in Northeastern Utah and three of the most important, relating to their motion characteristics imposed on Cache County and the surrounding area, are the East and West Cache faults, and the Northern Wasatch Fault shown on Figure 3-14.

The East Cache Fault is a north-south trending fault located along the eastern extremity of the Cache Valley, extending from a mile or more south of Avon, Utah, to well into Southern Idaho. The length

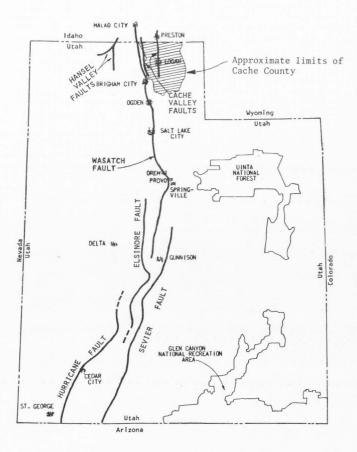


Figure 3-14. Locations of the major active faults in Utah by Cluff, Glass, Brogan (1974).

of this fault is about 70 miles (113 kilometers). Many scarps are visible along its length where it displaces the ancient Lake Bonneville sediments. The maximum credible earthquake for this fault is estimated to have a Richter magnitude of about 7.7 (from Bonilla, 1967, as presented by Seed, Idriss, and Kiefer, 1969).

The West Cache Fault, located generally along the western extremity of the Cache Valley is estimated to be about 55 miles (88 kilometers) in length and trends north-northwest. The maximum credible earthquake for this fault is estimated to be a magnitude of 7.5 (after Bonilla, 1967, as presented by Seed, Idriss and Kiefer, 1969).

The Northern Wasatch Fault is located along the western base of the Wasatch Mountain Range. It extends from near Gunnison, Utah, to Malad, Idaho, a length of about 215 miles (346 kilometers). Movement along this fault is more easily discerned than along the Cache Valley Faults and may indicate more recent movement (Cluff, Glass, and Brogan, 1974). This fault trends north-south and the maximum credible earthquake for this fault is magnitude 8.4 (after Bonilla, 1967, as presented by Seed, Idriss, and Kiefer, 1969). The energy that would be released from the maximum credible earthquakes along these three faults is shown on Figure 3-1.

A critical earthquake hazard exists in the State of Utah and presently, public officials are only beginning to perceive its seriousness. The close proximity of the majority of the population of Utah to major fault zones, particularly along the Wasatch Front, creastes a serious potential for not only an economic catastrophe, but also a tremendous loss of life. Approximately 85 percent of Utah's population is located within 5 miles (8 kilometers) of the Wasatch Fault (Kaliser, 1957). Additionally, nearly 100 percent of the population of Cache County is located within 5 miles (8 kilometers) of the East or West Cache Faults. In an effort to reduce the earthquake hazard, the general session of the 1977 legislature created an 11-member Seismic Safety Advisory Council. The purpose of the council is to recommend safety programs, review and recommend changes in current standards, promote new standards, and propose legislation.

The Utah Department of Natural Resources, Urban and Engineering Geology Section of the Utah Geological and Mineral Survey listed the following items concerning the earthquake hazard situation in Utah (U.S. Senate, 1975).

- Utah's citizens have experienced earthquakes virtually since settlement.
- At least 40 damaging earthquakes have occurred in the 128-year history since settlement.
- Scores of earthquakes are recorded in Utah each year.
- The earthquakes have generally shallow foci (points of origin within the earth), making them potentially quite destructive.
- Utah is traversed by many active fault zones.
- Approximately 41% of the state is in seismic zone #3 (major destructive damage may occur); 22% is in seismic zone #2 (expected moderate damage).
- There is extensive population density and economic development within the active fault zone in the state.

- Severe ground shaking is likely to be experienced in the state's metropolitan areas from future earthquakes.
- Destruction from one earthquake along in Utah, the August 1962 Cache Valley earthquake of Richter magnitude 5.7, amounted to \$1,700,000 (1974 dollars).
- Recent geologic movements along extensive fault breaks is in evidence. Two feet of vertical displacement occurred north of Great Salt Lake in 1934.

The possibility of death and total destruction such as that which has recently occurred in South America, Europe, and Asia is not likely in the United States because of better construction materials and techniques. However, the damage potential of an earthquake in Utah is as great or greater than the damage potential in California and Alaska. The potential in Utah may even be more serious due to the fact that such a high percentage of the population is located in close proximity to the active fault zones. Furthermore, there is a significant amount of unreinforced masonry construction and earthquake vulnerable construction. The intermountain seismic zone has generated large earthquakes in the historic past, and it can be expected by generate major earthquakes in the future.

CHAPTER 4

EVALUATION OF STRUCTURES

During this study, over 100 facilities were investigated from a geotechnical viewpoint. Many of these facilities were constructed with inadequate geotechnical considerations and consequently, are now subject to some degree of geologic hazard. In many cases, no evidence was found of geotechnical involvement in projects of vital public importance and the siting criteria was established by non-technical personnel.

Method of Evaluation and Limitations

As discussed previously, each structure included in the investigation was studied to determine the present exterior conditions and if possible, to identify any problems. Plans, specifications, design or construction notes, and reports on the facility were reviewed when available. In many cases, especially in the study of older structures, little or no information or documentation was available to indicate the design considerations. In these cases information was obtained from persons who were knowledgeable about the design, construction, or maintenance of the structure when possible. When information regarding a facility was not available, no judgement could be made concerning the level of geotechnical input other than by noting obvious deficiencies as assessed by examining the exterior of the structure and studying its geologic environment. Cooperation and accurate reporting of information about a facility from knowledgeable persons was vitally important for proper evaluation in this study. In some cases, the owners, maintenance personnel, and design professionals were very reluctant to cooperate and provide meaningful information. The effect of this reluctance may be that the evaluations made on the facility reflect less geotechnical input to design and construction than may actually have been provided.

As stated previously, the geotechnical evaluations presented in this study are based on present day standards for design and construction. Many significant geotechnical engineering advances have been made in recent decades and accordingly, the design standards for most facilities constructed in recent years are more rigorous than the standards used in the construction of older facilities. However, geologic hazards are not mitigated by age of the structure; therefore, in order to evaluate the current hazard exposure the study was based on present day standards.

The study involved the consideration of a wide variety of influencing factors. For example, the construction of certain facilities sometimes require special considerations which other structures may not need. The type and extent of the investigation necessary is governed by the type of structure, its use, the risk involved, and its association with similar facilities. For critical facilities, the service which is provided must not only be reliable during normal periods of use, but also during emergency situations or during the time of need.

Suggested guidelines to be followed when conducting geologic and geotechnical investigations have been adopted by many states and

Federal Agencies, by private design professionals, and by universities. Duke (1975), Slosson and Amimoto (1974), Torgerson (unverified date), Huber (1970), California Division of Mines and Geology (1973), and Ford (1977) are a few of the references available which provide guidelines for hazard investigations.

Many facilities of public importance were studied during this investigation. In order to illustrate the nature of some of the hazards, the facilities were divided into several categories such as schools, culinary water tanks, dams, etc.

Each category of structure is followed by a discussion of hazards of a general nature which may affect it. Examples of facilities exposed to geologic hazards are discussed individually in order to help illustrate the nature of the hazards. The discussion of individual facilities is also intended to indicate the types of future investigation necessary to mitigate the hazards.

Culinary Water Storage Tanks

General

A total of 34 culinary water storage reservoirs were studied. Nearly all of the water tanks serving Cache Valley communities are circular in shape and constructed of reinforced concrete. Many of the tanks are either entirely or partly buried. The foundations for these tanks are typically circular perimeter footings supporting the exterior walls with interior column footings supporting reinforced concrete columns. Reinforced concrete roof slabs are supported by the walls and columns. The capacities of these water storage tanks range from about 20,000 gallons (76 cubic meters) to 1,000,000 gallons (3800 cubic meters).

Most of the county's communities are located on the foothill benches and nearly all derive at least a portion of their potable water from the surrounding mountains. In order to utilize gravity flow to maintain a desirable water pressure for public use, the ideal locations for storage tanks are generally found on the highest benches near the base of the mountains. Unfortunately, major fault zones are also located near the mountain bases, thereby exposing many culinary water tanks and water distribution lines to an inherent geologic hazard. At least seven culinary water tanks are apparently located very near or within the fault zones of the East Cache or the West Cache Faults.

Unfortunately, many recently constructed water storage tanks in Cache Valley are exposed to certain geologic hazards, especially seismic hazards. Water storage facilities are structures which are critically important in the event of a major earthquake. The San Francisco Earthquake of 1906 and the fires that resulted, emphasize the need for water supply systems that are capable of providing an adequate volume of water under acceptable pressure both during and after an earthquake. Also, Kaliser (U.S. Senate, 1975, pp. 254) points out that each of the intergral parts of a city's water system deserves individual consideration as to its placement. Then, if any of several fault zones move, the reservoir and arterial system can still function

adequately. The same faults which constitute hazards to some of the water tanks in Cache Valley show displacement and active movements of many feet in some locations.

Other storage tanks examined during this study may also be located near fault zones, however, without a subsurface investigation such as trenching, the relative position of the faults cannot be easily identified.

Nearly every community of the valley is served by waterlines that cross the active fault zones. Measures can and should be taken to mitigate the danger of disruption of these lines due to fault movement. The fault zones could perhaps be easily identified when the trenches are excavated to lay new water transmission lines. Currently, however, no program is underway to identify or correct any hazard of this type.

Hyrum culinary water storage tank

In 1973, Hyrum constructed a new 1,000,000 gallon (3800 cubic meter) storage facility. This structure is located adjacent to an older tank with a capacity of about 260,000 gallons (980 cubic meters). Both water reservoirs appear to be founded on the steep face of a fault scarp. This is evidenced by the linear fashion of the scarp to the North and South and the abrupt change from unconsolidated soil deposits immediately west of the site, to rock outcrops immediately east of the site.

These two water storage facilities are also subject to erosion and possible landslide hazards. The soil surrounding the tanks is a fine, yellow-brown sandy silt. This soil is highly erodable and

the deposit is cut by numerous erosion gullies. Soils surrounding the old reservoir tank have become stabilized over a period of years by the accumulation of vegetation growth; however, two six-foot deep gullies were noted directly beside the new tank. Several deep, narrow erosional gullies which extend up to and possibly under the tank were noted.

Minor mud flows have occurred on the steep (1.3 horizontal to 1 vertical) construction cut slope above the tank. Continued unchecked erosion of this face may constitute a problem to the new tank. Training ditches located on each side of the construction slope have helped check the erosion of this area.

Cracks were noted in the west face of the new tank. The cracks are predominantly vertical and horizontal and spaced three to four feet (0.9 to 1.2 meters) apart. Leakage from these cracks was very minor. The precipitation of calcium carbonate from the water in the tank, has nearly sealed the fissures. The design engineer for this facility indicated that the cracks were probably caused from shrinkage of the concrete upon drying. Consideration was given to the possibility that water from the tank may be leaking out of cracks not observable externally. Several very narrow erosional gullies which were several feet deep were found to extend directly from the side of the tank and one area was found to be quite damp. The assumption that leakage of water may be causing the erosion is probably not valid because of the large quantity of water that would be necessary to form the gullies. It may be concluded, however, that at least some minor leakage is occurring underground as suggested by the cracks above the ground level.

Regardless of the source of the water, the erosion is a serious problem and should be checked to avoid future problems.

Cracking of the soil and signs of the initial stages of a landslide were observed 80 feet (±) (24 meters) below the tank along the side of the access road. This was occurring on a very steep cut slope of about 1.3 horizontal to 1 vertical.

Due to the nature of some of the problems identified above, an immediate inspection of the facility was recommended to the Hyrum City personnel. The site of this water tank presents special geologic and geotechnical problems. Although many different criteria must be considered in selecting a site for a given facility, it must be recognized that locating an important public facility such as this on an active fault scarp could present very serious problems. The degree of risk from possible movement along a known active fault should be thoroughly considered during the early planning stages of a project. A safer alternative location for this facility could probably have been selected.

Millville culinary water storage tank

Millville constructed a new 300,000 gallon (1140 cubic meter) reservoir in 1976. This structure is located approximately 200 feet (61 meters) south of an old 72,000 gallon (274 cubic meter) storage tank. Both facilities are located near the East Cache Fault zone. Several minor fault traces were recorded within the excavation for the tank and in a trench for one of the distribution pipelines. These were only minor traces and indicated slippage of probably less than one inch. A study of the landform and aerial photographs suggests

the site is in close proximity to the major active zone of slippage. The inherent danger of locating the tank in a fault zone was probably not studied adequately for this vital public facility.

There is no evidence that trenching or other subsurface investigation was conducted prior to siting and design of the facility even though the facility is located near a known geologically hazardous area (the East Cache Fault). The Millville City Mayor and the design engineer were notified about the minor fault traces at a time when the hazard could have been easily investigated further. The significance of the selected location, possible very close to a zone of major slippage, was apparently not realized and the facility was constructed with no additional investigation.

Providence culinary water storage tank

Similar fault hazards exist near two Providence City storage tanks. A new 1,000,000 gallon (3800 cubic meter) tank was constructed in 1976 adjacent to an old 130,000 gallon (494 cubic meter) tank at the entrance to Providence Canyon. A large sag depression and many linear fault scarps extending a few hundred feet north and south of the tank attest to the presence of active faulting very close to or possibly at the location of the tank. Rock outcrops suggest that the tank location may be slightly to the east of a major zone of slippage; however, a subsurface investigation such as trenching would be required to confirm the fault location.

An additional geologic hazard may exist from flooding. Both storage tanks are located in the flood plain produced by the incision of Providence Creek into the ancient Lake Bonneville deltaic deposits.

It was not determined if a study of the flooding potential was undertaken.

Wellsville culinary water storage tank

This 600,000 gallon (2280 cubic meter) tank was constructed in 1974 and is located in a very hazardous fault zone. Faults are located near two sides of the tank. One is located about 150 yards (137 meters) below the tank and the other is located about 80 feet (24 meters) (\pm) above the tank. Both fault traces exhibit large linear sag depressions several hundred feet long and 20 to 25 feet (6 to 8 meters) deep. A spring is located along the lower fault trace and may be associated with the fault. Some continuity can be seen between these faults and other possible scarp features farther to the north and south.

Culinary Water Supply Wells

Many of the valley communities supplement spring water supplies with drilled wells. The wells usually pump directly into the water distribution systems and may only operate during the months when the additional water is needed most. The water rights to many springs are shared jointly by municipalities and irrigation companies. In some cases, more than the legal share of water is drawn from the springs for public use which is then replaced by an equal quantity of water pumped directly into the canals from shallow wells.

The culinary water supply wells are essentially free of problems. The Bear River Health Department usually requires approval of the design for a proposed well. This usually includes the drillers log of the boring, the proposed locations of perforations, and the zones which will be sealed to prevent contaminated surface water from entering the system. The design is usually executed by the driller. Hyrum City reported that the infiltration of sand in one of their wells is a minor problem.

Culinary Water Supply Springs

General

Sixteen culinary water supply springs were evaluated during this study. Most of the springs have been in use for many years. The water rights are generally well established and most communities have made minor improvements periodically. The sources of water for many of the springs have not been studied and can only be inferred from knowledge of the geologic structure of the area. Contamination and apparent loss of water has prompted investigations such as dye testing in a few cases.

Most of the collection systems incorporate buried perforated lateral pipes which lead to collector boxes. From there, the water is usually gravity fed or pumped to storage reservoirs. The areas around the collector systems are generally fenced to prevent the infiltration of surface contaminants. In some cases, fences are not maintained and/or are not adequate.

Currently, a few of the culinary water supply springs probably do not meet minimum public health standards because of obvious contamination problems. The Environmental Protection Agency has recently (1977) adopted regulations which requires that any public water system meet minimum public health standards.

Providence culinary water supply spring

Providence obtains its water from a spring located approximately 2 miles (3.2 kilometers) above the entrance to Providence Canyon. A possibility of contamination of this spring exists because of a large open pit limestone mining operation located about 1 mile (1.6 kilometers) upstream. Dye tests indicate that the source of water from this spring originates above the mine and that the dye released at the mine reaches the spring very rapidly through groundwater flow channels. Currently, it appears that no particular measures are being taken to avoid the contamination hazard at the mine. At the time of this investigation the spring area was fenced; however, livestock had recently been grazing near the lateral collector pipes inside the fence.

A county ordinance adopted in 1977 states that no point sources of groundwater pollution may be discharged within a 1500 foot (457 meter) distance above a culinary water supply (springs and shallow wells). These pollution sources include septic tank leach fields, drainage from livestock grazing areas, or drainage from any other sources of pollution. This distance was arbitrarily chosen and is not always conservative. The Providence spring provides an example of a situation where contamination of the water from a source approximately 1 mile (1.6 kilometers) above the spring can readily contaminate the culinary water supply. This example illustrates that an individual evaluation of each proposed water supply development by a competent professional is necessary to ensure greater safety from contamination.

Cove culinary water supply spring

North Cove has had a long record of contamination problems with their water supply. The spring areas that supply the town's water are not adequately fenced and it was noted that cattle were allowed to graze directly within the water collection area.

Dams and Reservoirs

A total of ten dams were evaluated. At the present time, only five of these dams are considered as potentially dangerous to life and property in the event that they should fail. These are Porcupine, Hyrum, Logan First Dam, Newton, and Wellsville Dams.

Porcupine Dam

The largest dam in Cache Valley is Porcupine which is an earthfill dam. It is approximately 160 feet (49 meters) in height at the maximum section and retains 12,800 (15,788,600 cubic meters) of water. It is located on the East Fork of the Little Bear River and the town of Avon, the nearest community, is located about $3\frac{1}{2}$ miles (5.6 kilometers) downstream.

The dam was constructed in 1961 by the Utah Power and Water Board. Since its construction, several problems which are summarized below, have developed:

- Hydrostatic pressure caused the collapse of the original reinforced concrete retaining wall which served as a crest for the spillway inlet structure.
- Originally the spillway chute was constructed in bedrock. After the collapse of the inlet structure, it became apparent that because of erosion of the channel, it would need to be lined.

- Severe vibrations and cavitation of the outlet butterfly valve required that modifications be made.
- A series of leaks developed in the rock abutment on the south side of the dam. The leaks emerged from the lower part of the spillway at several locations.
- A horizontal pervious zone appeared in the embankment about 75 feet (23 meters) below the spillway crest. The downstream face was saturated until the reservoir was later drawn down to make the necessary repairs.
- At a distance of approximately 1/3 the way across the top of the dam from the left abutment, a slide developed near the top of the upstream slope. The slide was approximately 40 feet (12 meters) wide, 10 feet (3 meters) deep, and was estimated to be 50 or 60 feet (15 to 18 meters) long as measured down the upstream slope.
- Piezometers which were installed in 1962 need repair work in order to be operational. Vandals have broken security locks and reportedly dropped rocks down some wells.
- An area of incompetent rock located near the inlet to the spillway is currently undermining a 100 foot (±) (30 meters) high rock cliff. The overhanging rock mass is jointed and fractured and if it should collapse, is capable of significantly damaging and possibly plugging the spillway inlet structure.

Repairs to the spillway included the design and construction of a lined tunnel spillway. A design using a square, reinforced concrete culvert was selected. Repairs to the spillway have also included reconstruction of the reinforced concrete retaining wall which serves as a crest for the spillway inlet structure. Weep holes were provided to reduce hydrostatic pressure on the wall, rock bolts were used to anchor the south wall, and three steel pipe struts were installed to span across the inlet to provide additional lateral support for the wall. Also, pipe drains were installed along either side of the spillway to collect and drain seepage water from the spillway area.

The butterfly valve previously mentioned was moved to the extreme downstream end of the access tunnel serving the outlet pipeline. This was done to enhance the safety of the dam even if the vibration and cavitation problems with the valve were not eliminated. By providing a specially fabricated air intake, which allowed the introduction of large quantities of air into the pipeline, the vibration and cavitation problems were completely eliminated.

The leaks in the south abutment were sealed by grouting. The foundation was grouted to a depth of 160 feet (49 meters) forming a continuous curtain which tied into the existing grout curtain along the centerline of the dam. In order to help provide a water-tight condition, grout was also pumped under the spillway inlet structure. The grouting program was considered successful. The flow of water through the abutments was decreased from about 2 cubic feet per second (0.06 cubic meters per second) to about 1 cubic foot per second (0.03 cubic meters per second) (Palmer, 1964). Also, piezometers installed in the rock abutment, provided information that indicated the hydrostatic pressure would not be hazardous to the abutment.

In order to alleviate the seepage through the dam, a grout curtain was constructed. This grout curtain was constructed at approximately 75 feet (23 meters) upstream from the centerline of the dam and between the 40 and 90 foot (12 and 27 meter) depths below the dam crest. A horizontal, pervious, clayey gravel zone existing downstream from the grout curtain, was used as a blanket

drain to carry seepage water to the downstream face where it is collected in drain pipes and carried away from the dam. The construction of the grout curtain did not seal off the seepage completely; however, it is believed that it is functioning properly and has reduced the quantity of seepage water by a factor of perhaps five or ten (Palmer, 1964). Piezometers were installed and inspection revealed that the hazardous condition was checked.

The existence of the slide on the face of the dam was reported by the state engineer several seasons ago. It has been alleged that the owners were notified of the condition. No action had apparently been taken yet at the time of this investigation.

The loss of the piezometers which were installed in 1962 could be detrimental to the proper monitoring of the structure. These piezometers can provide valuable information regarding the seriousness of future stability problems which may arise for unknown reasons. Very little effort would be required to repair and protect these installations.

The damage caused by a rockfall in the vicinity of the spillway could be very significant. Sloughing has occurred over the past several years and continued undermining of the rock cliff presents a very real and present danger. An earthquake shock could possibly trigger such a collapse.

Hyrum Dam

This dam was constructed in 1934 by the U.S. Bureau of Reclamation. The dam is located at Hyrum and the community of Wellsville is situated approximately 3 miles (4.8 kilometers) downstream. The dam is of earthfill construction and is approximately 85 feet (26 meters) high at the maximum section. The impounded water is used primarily for irrigation and recreational purposes. The facility has performed well during its 40 years of service.

No particular geologic hazards effecting this structure were noted during this study. Hydrology studies have been conducted by the U.S. Bureau of Reclamation recently and the results have concluded that the present spillway is probably not adequate. Increasing the spillway capacity is currently being studied.

First Dam, Logan River

First Dam is a concrete gravity dam about 25 feet (8 meters) high. This is an old structure and is estimated to have been constructed in the early part of this century. The dam is located at the entrance to Logan Canyon. The reservoir capacity is roughly estimated to be less than 100 acre feet (123,000 cubic meters). Its capacity has been significantly reduced by siltation behind the dam over the years of operation.

Very little information was available concerning this structure. A field examination indicated that the dam is exposed to a fault hazard. Traces of slippage along the East Cache Fault can be observed short distances north and south of the dam. Any fault movement in this zone would result in almost certain damage or destruction of the dam and subsequent inundation of parts of the densely populated area directly downstream from the dam. The loose, saturated accumulation of silt within the reservoir would probably be transported by the flooding waters and cause additional damage. Geomorphic evidence also indicates the presence of a fault zone in the vicinity of the dam. The Logan River runs in a westerly direction a short distance upstream from the dam. Downstream from the dam, the river also follows a westerly course, but is oriented north-south along the East Cache Fault zone for several hundred yards, directly upstream and downstream from the dam. The sharp offset in the river channel is probably the result of erosion along the fault zone.

A highway cut several hundred yards north of the dam revealed approximately 12 feet (3.7 meters) of vertical displacement along the fault and aerial photographs (Cluff, Glass, and Brogan, 1974) indicated the extension of the fault to the north and south. Although no fault movements have occurred along this zone in recent history, the displacement revealed in the highway excavation suggests that the potential for surface rupture in the future is high.

Newton Dam

Newton Dam was constructed in 1946 by the U.S. Bureau of Reclamation. This dam is a zoned earthfill structure and is approximately 101 feet (31 meters) in height at the maximum section. The reservoir capacity is 5500 acre feet (6,780,000 cubic meters). The dam is located on Newton Creek, $3\frac{1}{2}$ miles (5.6 kilometers) north of Newton. The primary function of the facility is to impound irrigation and recreational water.

This study identified no geologic hazards influencing this facility. The right abutment of the dam is founded on an ancient landslide which occurred on the west side of Little Mountain, however,

the slide shows no evidence of any recent geologic activity resulting from construction of the dam.

Wellsville Dam

This dam is located within the Wellsville City limits. It is a very old earthfill structure and very little is known about it. The dam is approximately 20 feet (6 meters) high at the maximum section, 300 feet (91 meters) in length along the crest and about 10 feet (3 meters) wide at the top. The side slopes were roughly measured with a hand held inclinometer and found to be between 1.1:1 to 1.4:1 (horizontal: 1 vertical). These side slopes are considered to be very steep but no sign of instability is apparent. Several wet areas below the dam may be attributed to minor seepage through the base or under the dam. Heavy vegetative cover has grown on the side slopes of the embankment. The freeboard was measured and found to be approximately l_2^1 feet (0.46 meters) below the crest elevation. It is controlled by an overflow spillway on the left abutment of the dam. The reservoir capacity is unknown but roughly estimated to be less than 50 acre feet (61,700 cubic meters).

Schools

Two high schools, 3 junior high schools and 14 elementary schools, in the Cache County and Logan School Districts were reviewed during this study. Very little information was available to assess most of the schools; hence, the information collected was based largely on field observations and on reports from persons with knowledge of history of the structures.

Perhaps the most critical hazard to which the schools are exposed, is that of ground shaking from earthquakes. Many of the older structures are showing some signs of deterioration and distress. While these schools may be safe with only static forces acting on them, dynamic forces induced by an earthquake may exceed their reserve strength and cause extensive damage. Problems stemming from the August 1962 Cache Valley Earthquake serves as an appropriate example of the hazards that may exist. During this earthquake, ominous cracks developed in the Lewiston Elementary School, Park Elementary School in Richmond, and the old Logan Junior High School. The old Logan Junior High School was later razed as a result of this damage. Other damage to the schools included window breakage at some schools, cracking of the smoke stack at the Logan High School, and destruction and collapse of a portion of the stone capping on the North Cache Junior High School at Richmond. After the earthquake, steel tie rods were placed in the Lewiston Elementary School to help reinforce the structure.

The magnitude of the August 1962 earthquake was rated as 5.7 on the Richter scale; however, an earthquake of much greater magnitude is possible in the Cache Valley. Experience suggests that great damage can be expected from an earthquake of magnitude greater than the 1962 Cache Valley event.

The hazard exposure due to locating schools in fault zones is another important consideration. Although Cache County schools are not currently faced with this hazard exposure, it was found that site selection processes of the past have not considered proximity of the site to fault zones.

Four exploratory drill holes were bored at the site of the Sky View High School at Smithfield in 1962. An examination of the boring logs suggests that the liquefaction potential of the underlying soil should be studied. The density of underlying fine, silty sand layers is quite low as indicated by standard penetration tests. The height of the static water table could not be determined from the boring records, however, if the water table is near the surface, a critical liquefaction hazard may exist.

The indication that liquefaction could be a hazard is important since many other areas, especially along the Bear River, are probably underlain by similar soils. During the August 1962 Cache Valley Earthquake, the development of sand boils in several areas along the Bear River indicated that a condition of initial liquefaction had occurred. Earthquakes with greater magnitudes and longer durations may cause extensive liquefaction in the locations that displayed the characteristics of initial liquefaction.

CHAPTER 5

CONCLUSIONS AND RECOMMENDATIONS

Purpose of Study

The purpose of this study was to review existing and proposed publicly used facilities in Cache County, Utah, with regard to geologic hazards and to study the present level of geologic and geotechnical input involved in the siting, design and construction of these structures. The study was intended to indicate the present general level of geologic hazard existing in Cache County. It was also intended to provide information that would help define the nature of problems stemming from these geologic hazards and to encourage the review and revision of currently inadequate public policies pertaining to the siting, design, and construction of critical public facilities.

Limitations

The cooperation from owners, designers, constructors, and anyone knowledgeable about individual facilities was vitally important for a complete and proper review of the facilities. Some or all of these individuals, in some cases, were reluctant to provide meaningful information. In other cases, especially those pertaining to many older structures, little or no information or documentation was available to indicate the design considerations or thoroughness of the geotechnical investigation. Therefore, it was frequently the case, that only the surficial geologic environment could be studied to indicate any problems. A knowledge of the subsurface conditions such as those provided by a proper geologic and geotechnical investigation, may indicate either a greater or lesser exposure to geologic hazards than can be deduced from surficial evidence.

The scope of this study did not permit investigations of the structural capability of facilities to withstand strong ground shaking motion. The study was limited to critical public facilities within Cache County. Facilities associated with Utah State University, however, were not investigated.

Conclusions

Earthquake hazards probably constitute the greatest geologic hazard threat to publicly used facilities in Cache County. Major damage related to strong ground shaking, surface fault rupture, liquefaction or miscellaneous ground failure may result from movement along any one of at least three major faults in or near Cache Valley. These faults and the maximum credible earthquake associated with each are indicated below:

- The East Cache Fault, having a north-south orientation and located along the eastern edge of Cache Valley, is estimated to be capable of producing a maximum credible earthquake magnitude of 7.7 on the Richter scale.
- The West Cache Fault, having a north-south orientation and located at the western margin of Cache Valley, is estimated to be capable of producing a maximum credible earthquake magnitude of 7.5 on the Richter scale.
- The Wasatch Fault, having a north-south orientation, extending for over 200 miles (322 kilometers) and located about 6 miles

(9.7 kilometers) west of Cache Valley, is estimated to be capable of producing a maximum credible earthquake magnitude of 8.4 on the Richter scale.

A large magnitude earthquake occurring along any of these three major faults would probably cause a considerable amount of damage to many of the structures reviewed during this study, especially many older structures.

Strong ground shaking resulting from earthquakes is considered to be the greatest seismic hazard affecting facilities. Other seismic hazards, such as fault rupture, may present critical problems to some structures. Many past earthquakes in the United States, in areas with similar construction to that of Cache Valley, and even the records of damage in recent Utah and Cache Valley earthquakes supply ample evidence to support the fact that strong ground shaking is usually the major cause of damage. Kaliser (U.S. Senate, 1975, pp. 242) reports that the ground shaking damage from the August 1962 Cache Valley Earthquake alone was \$1,700,000 (1974 dollars). This earthquake had a Richter magnitude of 5.7 and mainly affected areas of Logan, Smithfield, Richmond and Lewiston.

Some facilities in Cache County are exposed to surface fault rupture hazards. If a structure is located astride a fault which displaces more than a few inches, extensive damage or total destruction is likely to occur. The surest way to avoid this hazard is to identify the fault zone and avoid construction there.

The active faults are located near the base of the mountain ranges in Cache County. Certain critical facilities such as culinary water supply storage tanks, because of economic considerations stemming from hydraulic design considerations, are also typically located on sites near the base of the mountain ranges. Thus, an inherent geologic hazard exists at virtually all of these otherwise desirable sites. This condition dictates the need for careful individual investigation of all sites in close proximity to known fault zones.

Major geologic hazards other than those related to earthquakes do not appear to affect most structures included in this study. Most of the county's communities have been developed on recessional benches produced by ancient Lake Bonneville as it receded. The soil types found in these benches, especially the higher benches, is generally dense sand and gravel and is usually well suited for construction. However, some problems such as erosion and the development of minor landslides have occurred at or near some of the facilities studied in the bench areas. These problems have generally developed in areas possessing steep slopes and/or excessive groundwater.

Although the structures reviewed do not indicate problems arising from subsidence due to compressible clay strata, many areas of the valley (generally, those below elevation 4500 feet) are underlain by such soils. Since the construction of major structures in these areas has been minimal, problems have usually not been encountered. The soft clays of the valley are highly compressible and extensive groundwater mining could lead to land subsidence problems. This hazard should be thoroughly investigated before extensive groundwater use is allowed.

Many critical facilities were found to be subjected to some degree of geologic hazard during this study. In an effort to emphasize

the critical need for adequate geologic and geotechnical investigations and analysis for important public structures, some of the hazards that threaten existing structures have been identified in Table 5-1. It is hoped that this identification will illustrate the nature of the hazards and indicate where additional future investigations are necessary in order to mitigate the hazards.

It was not possible, within the scope of this study, to properly assess the structural hazard resulting from strong ground shaking. Based on the reported damage from past Utah earthquakes and the reported damage from past earthquakes in other areas where construction techniques were similar, the hazard is considered to be high. This is especially true for many older structures.

General Recommendations

The "state-of-the-art" in earthquake engineering has advanced significantly in the last decade. It has come about largely as a result of the impact of several catastrophic earthquakes in California. Duke (1975) describes the impact of these earthquakes as follows:

- San Francisco, 1906 -- led to improvements in water supply systems, principally for fighting fire. The importance of system redundancy was emphasized.
- Long Beach, 1933 -- resulted in general adoption of lateral force provisions in California building codes. Several California electrical utilities adopted a seismic design criteria for important facilities which exceeded those required by local building codes.
- Kern County, 1952 -- electric utilities improved criteria for anchoring and bracing electrical equipment. Some earthquake resistive criteria were developed for tanks located on the ground and the importance of flexibility in connected piping was emphasized.

Table 5-1. List of structures with hazardous conditions.

Facility	Location	Hazard
Culinary Water Storage Tanks Hyrum	At mount of Blacksmith Fork Canyon	1. Astride major fault scarp 2. Minor landslide below tank 3. Erosive soil
Millville	Extreme S.E. corner of Millville	Very near or within major fault zone
Providence	At mouth of Providence Canyon	Very near or within major fault zone
Wellsville	Approximately 1 mile south of Wellsville	Located a few feet away from and between two major fault scarps
Culinary Water Supply Well Hyrum	East end of Main Street, Hyrum	Minor problem with sand infil- tration
Culinary Water Supply Springs Providence Canyon Spring	2 miles above entrance to Providence Canyon	Contamination of water supply from sources upstream

Table 5-1. Continued.

Facility	Location	Hazard
Cove Springs	Approximately 3/4 miles N.E. of town	Contamination of water supply from infiltration of surface pollutants
Dams and Reservoirs		
Porcupine Dam and Reservoir	Approximately 1½ miles east of the mouth of the East Fork of the Little Bear River	 Rockfall hazard near spill- way Loss of piezometer instru- mentation because of van- dalism Possible stability problems in embankment
Hyrum Dam	Southwest portion of Hyrum	Spillway capacity is possibly inadequate
First Dam Logan River	At mouth of Logan Canyon	 Located astride a major fault scarp Extensive populated area downstream
Schools Lewiston Elementary School	Lewiston	Cracked walls resulting from ground shaking during Cache Valley Earthquake, 1962

Table 5-1. Continued.

Facility	Location	Hazard
Park Elementary School	Richmond	Minor cracking resulting from ground shaking during Cache Valley Earthquake, 1962
01d Logan Junior High School	Logan	Structural damage resulting from ground shaking during Cache Valley Earthquake, 1962
North Cache Junior High School	Richmond	Window breakage, minor cracks and architectural damage
Logan High School	Logan	Window breaking, structural damage and near collapse of old smokestack
Sky View High School	Smithfield	Possible liquefaction of sub- surface soils

 San Fernando, 1971 -- this shock has had the greatest impact on the development of lifeline earthquake resistive design. Systems with and without aseismic design features were subjected to strong ground shaking and differential earth movements. The behavior of lifelines in this earthquake indicated some glaring hazards. (Duke, 1975)

California has benefited the most from these advances in the state-of-the-art. Other seismically active states have not yet begun to design aseismic capacity into many of their vitally important facilities. Many inexpensive remedies are available to help mitigate the earthquake hazards. However, a complete assessment of public facilities by the operating agency is essential just to determine and rank the hazards to which the facilities may be exposed, before mitigation measures can be initiated. By following the guidelines established by California agencies and by applying more stringent design requirements adapted for specific localities, structures can be made much more earthquake resistant.

As a result of the San Fernando event, the American Society of Civil Engineers was stimulated to organize the Technical Council on Lifeline Earthquake Engineering (TCLEE). It is their goal, "to elevate the level of engineering practice in design of lifelines to survive earthquakes". The council is expected to assume a role similar to that of the Structural Engineers Association of California which has been very instrumental in developing better design standards for that state.

Publically accepted levels of risk should be established for critical facilities in all communities. Legislation may be required to establish these risk levels or acceptable levels of performance. The development of aseismic design criteria should be guided by professional leadership and upgraded as needed when evaluated by field tests resulting from future earthquakes.

Operating agencies responsible for facilities should set strict guidelines and review procedures to assure the adequacy of assessments of the level of hazard exposure of their facilities. The Public Utilities Commission of California has set some general aseismic guidelines for seismic safety as reported by Duke (1975):

- Provide standby and storage facilities and alternate routes.
- Insure rapid restoration capability
- Provide interconnections with other utilities
- Meet and exceed standard building and safety codes
- Review and modify existing design criteria
- Review older structures
- Install instrumentation to record strong ground motion and monitor damage
- Plan major routes to avoid areas of known seismic hazard
- Recognize possible site amplifications in design criteria
- Coordinate emergency planning with other utilities and agencies (Duke, 1975).

The Earthquake Engineering Research Institute (EERI), is a national society composed of professional people from universities, industry associations, research organizations, professional societies, and many firms and agencies directly involved in the planning, design, construction, and operation of earthquake resistant structures. It is the role of EERI to help communicate and coordinate earthquake mitigation measures among the many organizations interested in the field. By subscribing to the principles established by EERI and becoming actively involved in formal self-education of the hazards, communities will be able to significantly reduce the earthquake hazard exposure of their facilities. Post-earthquake reconnaissance inspection forms for buildings and lifelines are available through EERI and will play an important role in collecting data to mitigate damage from future earth shocks.

Specific Recommendations

Development of public facilities in Cache County is currently proceeding with only minimal aseismic consideration. This should not be allowed to continue. Geologic and geotechnical investigations should be performed for all facilities of public importance. The degree of investigation required is dependent on the site location and should be determined by an individual who is experienced in the field. Geologic as well as geotechnical evaluation of sites should be mandatory whenever the construction takes place in the bench areas along either side of the valley or where surface fault rupture hazards have been identified (refer to Cluff, Glass and Brogan, 1974).

It is extremely important that geologic and geotechnical consideration be incorporated early in the planning stages for a facility. Proper identification and evaluation of hazards during the site selection phase of a project will generally save money in the long run. If a hazardous site is chosen unknowingly and the operating agency becomes committed to the location because of contractual or other reasons, or if problems develop after the construction of a facility is started, the associated costs to correct the problems are likely to be substantially greater than the increased cost associated with a detailed geologic and geotechnical investigation. It must be recognized, however, that only trained, experienced individuals may be capable of fully and correctly assessing the nature of the hazards as they exist.

When a facility must be located in a potentially vulnerable location, the investigation of the hazard should insure that the facility possesses a high order of dependability of performance. Construction should not take place astride active fault traces. However, when a site must, of necessity, be located within a fault zone, the investigation should define the exact location of surface fault ruptures if possible. Finally, a plan of action should be developed to quickly restore necessary services from damaged facilities.

Certain lifelines such as water lines, sewer lines, gas or petroleum lines, and power and communication lines must cross active faults. The location of the fault should be identified and if the competence of the fault cannot be guaranteed, the necessary materials for rapid repair should be provided where they are readily available. Additionally, it is recommended that the lines cross at or near the ground surface to facilitate emergency repair work. Automatic shut-off valves and flexible joints should be incorporated to control the

damage resulting from the disruption of the lifeline itself. Figure 5-1 (from Lew, Leyendecker and Dikkers, 1971) illustrates the damage and hazard exposure resulting from the rupture of gas, water, and sewer lines in the same location. The facilities were not only inoperable when needed most, but also a significant problem was created by the lifelines themselves.

Structural damage resulting from strong ground shaking has been identified as an extreme hazard during earthquakes. The Structural Engineers Association of California has been effective in reducing the hazard by the adoption and enforcement of stringent aseismic design criteria. The latest edition of the Uniform Building Code (1976) although significantly superior to previous editions, is regarded by many as providing only minimal earthquake resistive design criteria for structures. In seismically active areas, critical structures such as hospitals and emergency service structures, schools or other high occupancy structures, lifelines, and other important public facilities, may require more stringent design criteria than the criteria set forth by the Uniform Building Code (1976).

Many times minor structural details can help mitigate a significant portion of the total damage. Figures 5-2 through 5-5 from Lew, Leyendecker, and Dikkers (1971) provide glaring examples of the omission of such details. In Figure 5-2, telephone switching equipment at the Sylmar central office of General Telephone Company, is shown to be totally disrupted. If the columns of equipment had been tied together structurally, failure would not have occurred. This situation may be likened to that of rows of bookshelves in a

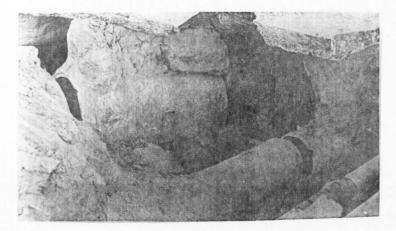


Figure 5-1. Ruptured sewer, water, and gas lines after the San Fernando Earthquake (after Lew, Leyendecker, and Dikkers, 1971).

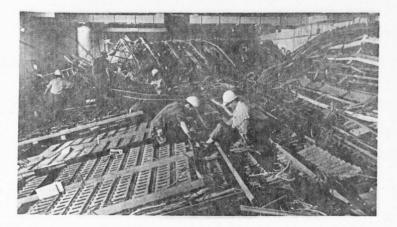


Figure 5-2. Toppled and damaged telephone switching equipment at Sylmar Central Office of General Telephone Company (after Lew, Leyendecker, and Dikkers, 1971).

library. If the rows are tied together to act as a unit rather than each row being supported individually, the unit will become much more stable. Instead, if the rows act individually, and one row topples, the result may be the same as a row of closely spaced dominos falling, each domino upsetting the next one.

Figure 5-3 shows an ambulance which was destroyed as a result of the collapse of a simple port structure. In this case, insufficient lateral support was provided for the relatively massive roof structure. The vertical columns were not capable of transferring the lateral force which was generated by the roof structure as a result of the earthquake. The emergency service vehicles parked in this port were rendered totally useless during the time when needed most.

Figures 5-4 and 5-5 show similar destruction. The electrical power transmission equipment shown in the figures was destroyed simply because of insufficient lateral support. A detailed evaluation of similar critical components of any lifeline system, should suggest many relatively inexpensive measures which can be taken to safeguard the system.

In the past, the governing society has been reluctant to accept the increased costs normally associated with hazard resistance analysis and design. Even in such seismically active areas as California, Alaska, and Japan, the incentive to provide earthquake resistant structures did not occur until after the occurance of major disasters. In Utah, a major disaster has not yet occurred in the densely developed areas. However, the potential exists. By initiating, at this time,

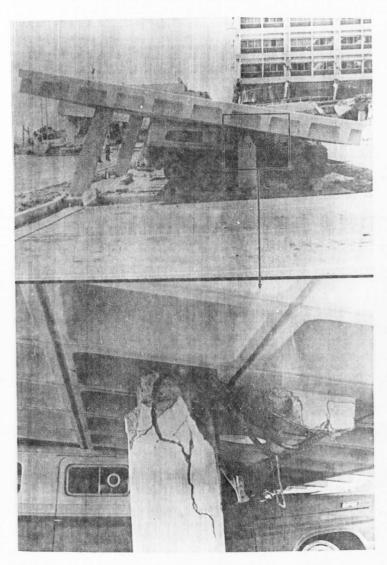


Figure 5-3. Destruction of emergency service vehicles parked in a collapsed ambulance port (after Lew, Leyendecker and Dikkers, 1971).

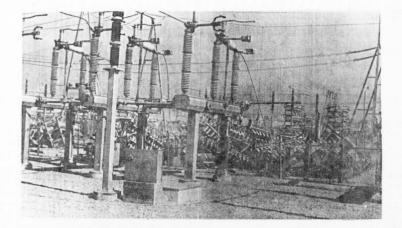


Figure 5-4. Destroyed electrical equipment at Sylmar Converter Station as a result of the San Fernando Earthquake (after Lew, Leyendecker, and Dikkers, 1971).

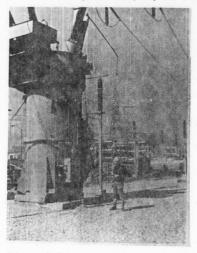


Figure 5-5. Destroyed electrical equipment at Sylmar Converter Station as a result of the San Fernando Earthquake (after Lew, Leyendecker, and Dikkers, 1971).

Programs to help educate the general public about existing geologic hazards, steps can be taken to mitigate a disaster before it occurs.

Providing hazard resistant structures is the responsibility of everyone. No single organization can be expected to carry the burden by itself. First of all, the government should establish a basis for an accepted level of risk to be associated with each type of structure used by the public. Guidelines should be established to help direct future development toward a high order of dependability of performance, and review procedures for existing vulnerable structures should be established.

The public should be responsible for requiring that future development and review of hazardous facilities provides for structures which function to protect life, property, and activities against geologic hazards and against a hazard from the facility itself. The public must realize that additional costs may be associated with earthquake resistant design and must be willing to accept this burden in exchange for safer structures.

The owner of a facility is responsible for providing for proper geologic and geotechnical investigations. The degree of the investigation required should be dependent on the site location and the type of facility. The degree of investigation should also fall within the guidelines as set by government agencies. The owner should obtain an investigation and design which is balanced between economy and risk and falls within the set guidelines.

A geotechnical engineer and/or geologists are responsible for informing everyone of the degree of subsurface investigation necessary

for a site. They should suggest to the owner when additional investigation is necessary. The architect and structural engineers are also responsible for requiring that competent personnel and proper subsurface investigations are provided for vital public facilities.

Finally, the constructor is reponsible for seeing that the facility is properly constructed as designed. Adherence to minor details during the construction of a facility can provide for a much safer structure.

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APPENDICES

Appendix A

The data in this appendix includes a list of 609 earthquakes which occurred in Utah during the period 1850 through June 1965

(from Cook and Smith, 1967)

Appendix A

KEY TO EARTHQUAKE DATA

Time of earthquake occurrence -- (year, month, day, hour, minute, second) given in Greenwich Civil Time (GCT)¹.

Earthquake location -- North latitudes (LAT-N) and West longitudes (LONG-W) are listed with the following accuracy: University of Utah data - nearest 0.01 degree; Coast and Geodetic Survey Data - nearest 0.1 degree; and Non-instrument recorded data geographical location, usually where greatest intensity was reported.

- Richter magnitude -- (MAG). Magnitudes of earthquakes occurring prior to 1950 were obtained by converting the intensities, (modified Mercalli scale) which were reported by Williams and Tapper (1953), to Richter magnitudes. Magnitudes of earthquakes occurring from 1950 through June 1962 were as listed by the Coast and Geodetic Survey. Magnitudes of earthquakes occurring from July 1962 through June 1965 were as listed by the University of Utah.
- Focal depth -- (DEP) given in kilometers, as listed by the Coast and Geodetic Survey. When the Coast and Geodetic Survey focal depths were not available, the depth was arbitrarily and approximately placed at 20 km.

¹Greenwich Civil Times are seven hours later than Mountain Standard Times.

Source of data -- (S) is listed as follows:

C -- Coast and Geodetic survey data

U -- University of Utah data

W -- Williams and Tapper (1953)

Epicenter determination -- (D) was made by both the Coast and Geodetic

survey and the University of Utah when the letter "D" was included. Remarks -- This column gives general information about each event.

1853	DEC 01	19 15	 	4.3	v	W	NEPHI WASATCH FAULT	
1853	DEC 01	19 45	 	4.3	V	w	PROVO WASATCH FAULT	
1557	AU3 23		 	3.7	IV	w	PAROWAN HURRICANE FAULT	
1865	OCT 17	10 30		3.1	III	W	EPHRAIM THOUS LAKE FAULT & SHOCKS	
1872	VAR 27	07 52		2.0	II	w	SALT LAKE CITY WASATCH FAULT	
1873	JUL 31	03 15	 	4.3	V	w	BEAVER HURRICANE FAULT DAMAGE	
1573	DEC 15	14 00		3.7	IV	W	BEAR LAKE VALLEY BEAR LAKE FAULT	
1873	DEC 27	03 00		3.7	IV	*	FARMINGTON SEVERE SHOCK WASATCH FLT	
1874	JUN 18	06 00		3.7	IV	W	SALT LAKE CITY	
1874		07 00		3.7	IV	w	FELT SALT LAKE CITY TO MIDWAY	
	JUN 18			4.9	VI	W	MT PLEASANT AND MOPONI 3 SHOCKS	
1875	MAR 22			4.9	¥ 1		SAN PETE VALLEY SEVERAL SHOCKS	
1875	MAR 24						BEAR LAKE VALLEY BEAR LAKE FAULT	
1875	APR 06					2	BEAR LAKE VALLEY BEAR LAKE FAULT	
1875	APR 06						BEAR LAKE VALLET HEAR EARE FADET	
1876	NOV 30	05 00		3.1	TIT	w	CEDAR CITY HURRICANE FAULT	
1876	DEC 30	05 15		3.1	TIT	w	RICHFIELD COVE CREEK TUSHAR FAULT	
1877	JAN 01			3.7	IV	W	RICHFIELD FELT 3,000 SO MI TUSHAR FLT	
1877	JAN 15	12 00		3.1	III	W	SEVIER CITY TUSHAR FAULT 2 SHOCKS	
1877	MAR 05	09 00		2.0	II	W	SALT LAKE CITY WASATCH FAULT	
1878	JUL 21	12 00		5.0	II	W	SALT LAKE CITY WASATCH FAULT	
1878	AUG 14			4.3	V	W	COVE CREEK HURRICANE FAULT	
1878	AUG 15	13 40	 	2.0	II	W	COVE CREEK HURRICANE FAULT	
1878	AUG 15	14 40	 	2.0	II	W	COVE CREEK HURRICANE FAULT	
1878	AUG 16	02 44	 	3.7	IV	W	COVE CREEK HURRICANE FAULT	
1878	AUG 21	12	 	3.1	III	*	NORTH OF SALT LAKE CITY WASATCH FLT	
1878	SEP 07	19	 	3.1	III	W	SALT LAKE CITY WASATCH FAULT	
1878	DEC 02		 	2.0	II	W	PANGUITCH SEVIER FLT SEVERAL SHOCKS	
1850	JUL 12	05 00	 	4.3	v	w	BOX ELDER CO PORTAGE 2 SHOCKS	
1880	SEP 16	06 27		3.7	IV	w	SALT LAKE CITY WASATCH FAULT	
1880	DEC 27			3.1	III	w	KELTON HANSEL VALLEY FLT 2 SHOCKS	
1881	MAR 26	02 15		3.1	III	w	HEBRON DAMAGE	
1881	AUG 04	04 30		3.1	III	W	BEAVER HURRICANE FAULT 5 SHOCKS	
1881	AUG 04	19 30				W	BEAVER HURRICANE FAULT	
1881	OCT 16	07 00		3.1	III	W	MOUNT PLEASANT THOUSAND LAKE FAULT	
1883	SEP 25	11 00		3.7	IV	W	SILVER CITY	
1003	SEF 23	11 00		5.1				
1883	NOV 04		 			w	MILLARD CO COVE CREEK	
1884	NOV 10	08 50		6.1	VIII	W	NORTHEASTERN UTAH SEVERE DAMAGE	
1884	NOV 10	09 10		2.0	II	W	BEAR LAKE VALLEY CRAWFORD MIN FAULT	
1884	NOV 10	09 19	 	2.0	II	W	BEAR LAKE VALLEY CRAWFORD MIN FAULT	
1884	NOV 10	09 48	 	2.0	II	W	BEAR LAKE VALLEY CRAWFORD MIN FAULT	
1884	NOV 10	10 45		2.0	IT	W	BEAR LAKE VALLEY CRAWFORD MTN FAULT	
1894	NOV 11	08 55		2.0	II	W	BEAR LAKE VALLEY CRAWFORD MIN FAULT	
1894	NOV 11	14 00		2.0	II	W	BEAR LAKE VALLEY CRAWFORD MIN FAULT	
1884	NOV 12	08 50		2.0	II	W	BEAR LAKE VALLEY CRAWFORD MIN FAULT	
1984	NOV 12	09 35		2.0	II	W	BEAR LAKE VALLEY CRAWFORD MTN FAULT	
1584	NOV 12	12 05		2.0	II	W	BEAR LAKE VALLEY CRAWFORD MIN FAULT	
1884	NOV 13	08 55		2.0	II	w	BEAR LAKE VALLEY CRAWFORD MTN FAULT	
1884	NOV 13	10 40		2.0	IT	w	BEAR LAKE VALLEY CRAWFORD MIN FAULT	
1884	DEC 08	10 40		3.1	TIT	W	OGDEN HANSEL VALLEY FLT MANY SHOCKS	
1895	SEP 05	03 35		3.1	TIT	W	KANAB HEAVY SHOCK SEVIER FAULT	
1835	OCT 26	05 10		3.1	III	W	MINERSVILLE 4 EVENTS MINERAL MTS FLT	
1895	OCT 26	08 00		3.1	TTT	ÿ	BEAVER CO FRISCO BEAVER MIN FAULT	
1835	OCT 26	ug a0	 	3.1	III	W	BEAVER CO FRISCO BEAVER MIN FAULT	
1895	DEC 17	01 00		3.7	TV	w	CIRCLEVILLE TUSHAR FAULT	
1885	DEC 17	03 20		3.1	III	ŵ	TEASDALE THOUSAND LAKE FAULT	
		15 30		5.5	VIT	ŵ.	KANAB SEVIER FAULT DAMAGE	
1897	DEC 05 DEC 07	15 30		3.7	IV	÷.	MANTI THOUSAND LAKE FAULT	
1859					VI		WASHINGTON CO HURRICANE FAULT DAMAGE	
1891								
	APR 20	13 55		4.9				
1891	APR 20 SEP 13	13 55	 	3.1	III	W	CEDAR CITY HURRICANE FAULT	
1893	APR 20 SEP 13 AUG 30	13 55 04 48 23 30	 	3.1 3.7	III	w	CEDAR CITY HURRICANE FAULT SNOWVILLE HANSEL VALLEY FAULT	
1893 1894	APR 20 SEP 13 AUG 30 JAN 08	13 55 04 48 23 30 18 00	 	3.1 3.7 4.3	III V V	***	CEDAR CITY HURRICANE FAULT SNOWVILLE HANSEL VALLEY FAULT FISH SPRINGS FISH SPRINGS FAULT	
1893 1894 1894	APR 20 SEP 13 AUG 30 JAN 08 FEB 05	13 55 04 48 23 30 18 00 03 30		3.1 3.7 4.3 3.7	III V IV	н	CEDAR CITY HURRICANE FAULT SNOWVILLE HANSEL VALLEY FAULT FISH SPRINGS FISH SPRINGS FAULT KANOSH HURRICANE FAULT	
1893 1894 1894 1894	APR 20 SEP 13 AUG 30 JAN 08 FEB 05 FEB 06	13 55 04 48 23 30 18 00 03 30 15 00	 	3.1 3.7 4.3 3.7 3.1	III N IV III	¥¥	CEDAR CITY HURRICANE FAULT SNOWVILLE HANSEL VALLEY FAULT FISH SPRINGS FISH SPRINGS FAULT KANOSH HURRICANE FAULT KANOSH HURRICANE FAULT	
1893 1894 1894	APR 20 SEP 13 AUG 30 JAN 08 FEB 05	13 55 04 48 23 30 18 00 03 30 15 00 22 50		3.1 3.7 4.3 3.7 3.1 4.3		223	CEDAR CITY HURRICANE FAULT SNOWYLLE HANSEL VALLEY FAULT FISH SPRINGS FISH SPRINGS FAULT KANOSH HURRICANE FAULT KANOSH HURRICANE FAULT OGDEN WASATCH FAULT DAMAGE	
1893 1894 1894 1894	APR 20 SEP 13 AUG 30 JAN 08 FEB 05 FEB 06	13 55 04 48 23 30 18 00 03 30 15 00 22 50 22 25		3.1 3.7 4.3 3.7 3.1 4.3 3.7		***	CEDAR CITY HUPRICANE FAULT SNOWYILLE HANSEL VALLEY FAULT FISH SPRINGS FISH SPRINGS FAULT KANOSH HURRICANE FAULT KANOSH HURRICANE FAULT GGDEN WASATCH FAULT DAVAGE MT PLEAFANT THOUS LAKE FAULT DAVAGE	
1893 1894 1894 1894 1894	APR 20 SEP 13 AUG 30 JAN 08 FEB 05 FEB 06 JUL 18	13 55 04 48 23 30 18 00 03 30 15 00 22 50		3.1 3.7 4.3 3.7 3.1 4.3 3.7 3.1 4.3 3.7 2.1	111 v 111 v 111 v 111 v 111	****	CEDAR CITY HURRICANE FAULT SNOWILLE HAMSEL VALLEY FAULT FISH SPRINGS FISH SPRINGS FAULT KANOSH HURRICANE FAULT KANOSH HURRICANE FAULT GODEN WASICH FAULT DAMAGE MT PLEASANT THOUS LAKE FAULT DAMAGE SAMPETE CO GUNNISON SEVIER FAULT	
1893 1894 1894 1894 1894 1895	APR 20 SEP 13 AUG 30 JAN 08 FEB 05 FEB 06 JUL 18 JUL 27	13 55 04 48 23 30 18 00 03 30 15 00 22 50 22 25		3.1 3.7 4.3 3.7 3.1 4.3 3.7 3.1 4.3 3.7 3.7	111 1 1 1 1 1 1 1 1 1 1 1 1	*****	CEDAR CITY HUPRICANE FAULT SNOWILLE HAMSEL VALLEY FAULT FISH SPRINGS FISH SPRINGS FAULT KANOSH HURRICANE FAULT KONTH HURRICANE FAULT NO AND HURRICANE FAULT MARKET HURRICANE FAULT SAMPETE CO GUNNISON SEVIER FAULT NEPHI WASATCH FAULT	
1893 1894 1894 1894 1894 1895 1895	APR 20 SEP 13 AUG 30 JAN 08 FEB 05 FEB 06 JUL 18 JUL 27 JUN 07	13 55 04 48 23 30 18 00 03 30 22 50 22 25 05 30		3.1 3.7 4.3 3.7 3.1 4.3 3.7 3.1 4.3 3.7 3.1 3.7 3.1	III IV IV IV IV IV IV IV IV III	******	CEDAR CITY HURRICANE FAULT SNOWILLE HAMSEL VALLEY FAULT FISH SPRINGS FISH SPRINGS FAULT KANOSH HURRICANE FAULT KANOSH HURRICANE FAULT GODEN WASATCH FAULT DANAGE MT PLEASANT THOUS LAKE FAULT DAVAGE SAMPETE CO GUNNISON SEVIER FAULT NEPHI WASATCH FAULT LOGAN EAST CACHE FAULT	
1893 1894 1894 1894 1894 1895 1895 1896 1896	APR 20 SEP 13 AUG 30 JAN 08 FEB 05 FEB 06 JUL 18 JUL 27 JUN 07 SEP 13	13 55 04 48 23 30 18 00 03 30 15 00 22 25 05 30 01 30		3.1 3.7 4.3 3.7 3.1 4.3 3.7 3.1 4.3 3.7 3.7	111 1 1 1 1 1 1 1 1 1 1 1 1	*****	CEDAR CITY HUPRICANE FAULT SNOWILLE HAMSEL VALLEY FAULT FISH SPRINGS FISH SPRINGS FAULT KANOSH HURRICANE FAULT KANOSH HURRICANE FAULT ON PLEASANGH FUNJSIANGE GANDETE CO GUNNISON SEVIER FAULT NEPHI WASITCH FAULT LOGAN EAST CACHE FAULT HANKSVILLE FELT J.000 50 MI	
1893 1894 1894 1894 1894 1895 1895 1896	APR 20 SEP 13 AUG 30 JAN 08 FEB 06 JUL 18 JUL 27 JUN 07 SEP 13 OCT 03	13 55		3.1 3.7 4.3 3.7 3.1 4.3 3.7 3.1 4.3 3.7 3.1 3.7 3.1	III IV IV IV IV IV IV IV IV III	******	CEDAR CITY HURRICANE FAULT SNOWILLE HAMSEL VALLEY FAULT FISH SPRINGS FISH SPRINGS FAULT KANOSH HURRICANE FAULT KANOSH HURRICANE FAULT GODEN WASATCH FAULT DANAGE MT PLEASANT THOUS LAKE FAULT DAVAGE SAMPETE CO GUNNISON SEVIER FAULT NEPHI WASATCH FAULT LOGAN EAST CACHE FAULT	

YEAR NO DAY HE MN SEC LAT-N LONG-W MAG INT DEP 5 D REMARKS

1898										
	FEB 21	00 1	00						W	CORINNE
1899	NOV 10	03 1	no				3.7	IV	W	BEAVER HURRICANE FAULT
1899	DEC 13						3.7	IV	w	SALT LAKE CITY WASATCH FAULT
1900	AUG 01						5.5	VII	W	EUREKA DISTRICT DAMAGE
1900	AUG 01						2.0	TT	W	EUREKA DISTRICT
							2.0	IT	W	EUREKA DISTRICT
1900	AUG 01									EUREKA DISTRICT
1900		16					2.0	II	W	
1901	AUG 11						3.1	III	w	SALT LAKE CITY WASATCH FAULT
1901	AUG 11	18 1	0.0				3.1	III	W	PROVO WASATCH FAULT DAMAGE
1901	NOV 14						2.0	II	W	RICHFIFLD SHOCKS NOV 14 TO NOV 30
1901	NOV 14	04	39				6.7	IX	W	RICHFIELD ABOUT 35 SHOCKS TUSHAR FLT
1902	JAN 05						3.1	III	W	BEAR LAKE VALLEY REAR LAKE FAULT
1902	JUN 01						3.1	TIT	w	BEAVER SEVERAL SHOCKS HURRICANE FLT
1902	JUL 31						4.3	V	W	BEAVER CO BEAVER HURRICANE FAULT
	NOV 17						6.1	VIII	*	PINE VALLEY 2 SHOCKS DAMAGE
1902							4.9	vr	~	PINE VALLEY MANY SHOCKS
1902	DEC 05							TIT		RICHFIELD TUSHAR FAULT
1903	JUL 12	16					3.1			OGDEN-SLC AREA WASATCH FAULT
1903	JUL 23						3.1	III	W	
1903	NOV 04	23 4	40				3.1	III	w	WASHINGTON CO ST GEOPGE
1903	NOV 23	10 1	00				4.3	V	w	WASHINGTON CO ST GEORGE
1905	NOV 11	23 1	0.0				4.3	V	W	SNOWVILLE HANSEL VALLEY FAULT
1906	FEB 21								2	SEVIER CO ELSINORE TUSHAR FAILT
1900	MAY 24						4.3	V	W	OGDEN 3 SHOCKS WASATCH FAULT
1908	APR 15						4.9	VI	*	MILFORD BEAVER MTS FLT 5 SHOCKS
	OCT 05						6.7	IX	w	NORTHWESTERN UTAH FELT 30,000 50 MI
1909							0.1			NORTHWESTERN UTAH HANSEL VALLEY FLT
1909	001 06							v	W	GARLAND HANSEL VALLEY FLT DAMAGE
1909	NOV 17						4.3			GARLAND MANY AFTERSHOCKS OCT TO DEC
1909	NOV 17						4.3	v	W	
1910	JAN 10						4.9	vt	W	ELSINOPE TUSHAR FAULT
1910	JAN 10	13	30				4.9	VI	W	ELSINORE TUSHAR FAULT
1910	JAN 10	13 4	45			******	4.9	VI	W	ELSINORE TUSHAR FAULT
1910	JAN 10	20 3	25				4.9	VT	W	ELSINORE TUSHAR FAULT
1910	JAN 1U						4.9	VI	W	ELSINORE TUSHAR FAULT
1910	JAN 11						4.9	VT	W	ELSINORE TUSHAR FAULT
1910	JAN 12						4.9	vr	W	ELSINORE TUSHAR FAULT DAMAGE
1910	DA4 15	0.5	10				4.5			LEGITORE TOURIST CONTRACT
							4.9	V!	w	ELSINORE TUSHAR FAULT
1910	JAN 12									SALT LAKE CITY 2 SHOCKS WASATCH FLT
1910	MAY 03						5.0	Ţ	W	SALT LAKE CITY WASATCH FAULT DAMAGE
1910	MAY 22						5.5	VII	14	
1910	MAY 22	15 3	33 .				5.5	VII	w	SALT LAKE CITY WASATCH FAULT
1910	MAY 22	18 2	26				5.5	VII	м.	SALT LAKE CITY WASATCH FAULT
1910	MAY 25	15 4	45				3.1	TIT	W	SALT LAKE CITY WASATCH FAULT
1910	MAY 25	0.5.1	15				2.0	II	W	SALT LAKE CITY WASATCH FAULT
1913	0CT 20									
				F 10 10 10			3.7		W	PANGUITCH SEVIER FAULT
							3.7	IV		PANGUITCH SEVIER FAULT WASATCH FRONT SALT LAKE TO OGDEN
1914	APR 03	16 1	nó (4.3	V	W	WASATCH FRONT SALT LAKE TO OGDEN
1914	MAY 13	16 1 17 1	06 15				4.3	VII	W	WASATCH FRONT SALT LAKE TO OCCEN N WASATCH FRONT WASATCH FAULT DAMAGE
1914	MAY 13 DEC 14	16 1 17 1 05 3	15 15				4.3	VII	ਵ ਵ	WASATCH FRONT SALT LAKE TO OGDEN N WASATCH FRONT WASATCH FAULT DAMAGE ENTERPRISE
1914 1914 1914	MAY 13 DEC 14 DEC 21	16 17 05 05	06 15 30 25				4.3 5.5 4.3 3.1	VII V III	X X X	WASATCH FRONT SALT LAKE TO OCCEN N WASATCH FRONT WASATCH FAULT DAMAGE ENTERPRISE HURRICANE, PINE VALLEY, AND PINTO
1914 1914 1914 1915	MAY 13 DEC 14 DEC 21 FEB 12	16 17 05 05 19	06 15 30 25				4.3 5.5 4.3 3.1 3.7	VII V III IV	ਵ ਵ	WASATCH FRONT SALT LAKE TO OCCEN N WASATCH FRONT WASATCH FAULT DAMAGE ENTERPRISE HURRICANE, PINE VALLEY, AND PINTO ENTERPRISE
1914 1914 1914 1915 1915	MAY 13 DEC 14 DEC 21 FEB 12 FEB 13	16 17 05 05 19 03	15 30 25 50				4.3 5.5 4.3 3.1 3.7 3.7	VII V III IV IV	* * * * *	WASATCH FRONT SALT LAKE TO OCCEN N WASATCH FRONT WASATCH FAULT DAWAGE ENTERPRISE HURRICANE, PINE VALLEY, AND PINTO ENTERPRISE ENTERPRISE
1914 1914 1914 1915	MAY 13 DEC 14 DEC 21 FEB 12	16 17 05 19 03 04	06 15 30 25 50 30 30	·			4.3 5.5 4.3 3.1 3.7 3.7 2.0	VII V III IV IV II	****	WASATCH FRONT SALT LAKE TO OCOEN N WASATCH FRONT WASATCH FAULT DAMAGE ENTERPRISE HURRICAME, PINE VALLEY, AND PINTO ENTERPRISE ENTERPRISE EMTERPRISE WERY JOES VALLEY FAULT
1914 1914 1914 1915 1915	MAY 13 DEC 14 DEC 21 FEB 12 FEB 13 APR 25 JUL 15	16 17 05 19 03 04 22	06 15 30 25 50 30 30 30				4.3 5.5 3.1 3.7 2.0 4.9	V V III IV IV II V V V	* * * * *	WASAICH FRONT SALT LAKE TO GOPEN N WASAICH FRONT WASAICH FAULT DAWAGE ENTERPRISE ENTERPRISE ENTERPRISE EWERV JOES VALLEY FAULT SPRINSULL-SLC WASAICH FAULT
1914 1914 1914 1915 1915 1915	MAY 13 DEC 14 DEC 21 FES 12 FES 13 APR 25	16 17 05 19 03 04 22	06 15 30 25 50 30 30 30	·			4.354.354.33.77093	VII VII IV IV VI VI VI VI VI	****	WASAICH FRONT SALT LAKE TO GOPEN N WASAICH FRONT WASAICH FAULT DAWAGE ENTERPRISE UNITERPRISE ENTERPRISE ENTERPRISE EVERY JOES VALLEY FAULT SPRINOVILLE-SLC WASAICH FAULT BEAR RIVER VALLEY MANSEL VALLEY FLT
1914 1914 1914 1915 1915 1915 1915	MAY 13 DEC 14 DEC 21 FEB 12 FEB 13 APR 26 JUL 15 JUL 30	16 17 05 19 03 04 22 19	06 15 30 25 50 30 30 50 50				4.35.53.17.7.094.31	V V III IV IV II V V V	****	WASAICH FRONT SALT LAKE TO OCPEN N WASAICH FRONT WASAICH FAULT DAWAGE ENTERPRISE UNREIGNE, PINE VALLEY, AND PINTO ENTERPRISE EWERV JOES VALLEY FAULT SPRINGVILLE-SLC WASAICH FAULT BEAR RIVER VALLEY HANSEL VALLEY FLT STANSRURY RANGE
1914 1914 1915 1915 1915 1915 1915 1915	MAY 13 DEC 14 DEC 21 FEB 12 FEB 13 APR 26 JUL 15 JUL 15 JUL 15	16 17 05 19 03 04 22 19	06 15 30 25 50 30 30 50 50				4.35.53.17.7.094.31	VII VII IV IV VI VI VI VI VI	****	WASATCH FRONT SALT LAKE TO OCPEN N WASATCH FRONT WASATCH FAULT DAVAGE ENTERPRISE ENTERPRISE ENTERPRISE ENTERPRISE ENTERPRISE ENTERPRISE PRIMILE-SLC WASATCH FAULT BEAR RIVER VALLEY HANSEL VALLEY FLT STANSHIPY RANGE THISTLE
1914 1914 1915 1915 1915 1915 1915 1915	MAY 13 DEC 14 DEC 21 FEB 12 FEB 13 APR 26 JUL 15 JUL 15 JUL 15 JUL 15 JUL 120 LUG 11 SEP 20	16 17 05 05 05 04 22 04 22 18 10 01	16 15 30 30 30 50 28				4.3 5.5 3.17 7.09 4.3 1.1 5.1	V VII V V V V V V V V V V V V V V V V V	****	WASAICH FRONT SALT LAKE TO GOPEN N AGASICH FRONT WASAICH FAULT DAWAGE ENTERPRISE ENTERPRISE ENTERPRISE ENTERPRISE VALLEY FAULT EPRIT DEAR RIVELE-SLC WASAICH FAULT STANSHURY RANGE THISTLE
1914 1914 1915 1915 1915 1915 1915 1915	MAY 13 DEC 14 DEC 21 FEB 12 FEB 12 FEB 13 APR 26 JUL 15 JUL 30 AUG 11 SEP 20 OCT 02	16 1 17 1 05 1 05 1 19 0 3 1 04 1 22 1 19 1 22 1 10 1 23 4	06 15 30 30 30 30 50 20 20 20 20				4.3 5.5 3.1 3.7 7 2.0 4.9 4.3 6.1 3.1 2.0	VII VI VI VI VI VI VI VI VI VI VI VI VI	*********	WASAICH FRONT SALT LAKE TO OCPEN N WASAICH FRONT WASAICH FAULT DAWAGE ENTERPRISE ENTERPRISE ENTERPRISE EMTERPRISE EMTERPRISE SPRINGUILLE-SLC WASAICH FAULT BTAR RINGR VALLEY HANSEL VALLEY FLT STATT RANGE SALT LAKE CITY WASAICH FAULT SALT LAKE CITY WASAICH FAULT
1914 1914 1915 1915 1915 1915 1915 1915	MAY 13 DEC 14 DEC 21 FEB 12 FEB 13 APR 25 JUL 15 JUL 30 4UG 11 SEP 20 OCT 02 OCT 03	16 17 05 19 03 04 22 04 22 04 10 23 01 23 01	06 15 30 30 30 30 50 20 29 41 50				4.3 5.5 3.1 3.7 2.0 4.9 4.3 6.1 3.1 2.0 3.1	VII VII VI VI VI VI VI VI VI VI VI VI VI	**********	WASAICH FRONT SALT LAKE TO OCPEN N WASAICH FRONT WASAICH FAULT DAWAGE ENTERPRISE ENTERPRISE ENTERPRISE EMTERPRISE EMTERPRISE SPRINGUILLE-SLC WASAICH FAULT BTAR RINGR VALLEY HANSEL VALLEY FLT STATT RANGE SALT LAKE CITY WASAICH FAULT SALT LAKE CITY WASAICH FAULT
1914 1914 1915 1915 1915 1915 1915 1915	MAY 13 DEC 14 DEC 21 FEB 12 FEB 13 APR 25 JUL 15 IUL 30 4UG 11 SEP 20 OCT 02 OCT 03	16 17 05 19 03 04 22 04 22 04 23 04 23 012 0 23 0 23	no 15 30 30 30 30 20 20 20 20 20 20 20 20 20 20 20 20 20				4.3 5.5 3.1 3.7 3.7 2.0 4.9 4.3 6.1 3.1 2.0 3.1 4.9	V III V V V V V V V V V V V V V V V V V	* * * * * * * * * * * * * * * * * * *	WASATCH FRONT SALT LAKE TO GOPEN N WASATCH FRONT WASATCH FRULT DWAGE ENTERPRISE ENTERPRISE ENTERPRISE ENTERPRISE ENTERPRISE ENTERPRISE ENTERPRISE ENTERPRISE ENTERPRISE ENTERPRISE ENTERPRISE ENTERPRISE ENTERPRISE ENTERPRISE ENTERPRISE ENTERPRISE ENTERPRISE ENTERPRISE MASATCH FRONT SALT LAKE CITY WASATCH FAULT SALT LAKE CITY WASATCH FAULT N UTAH MANSEL VALLEY FAULT
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LAT-N LONG-W MAG INT DEP 5 D REMARKS

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			000		white sectors		
1921	JUH 02	21 30	-		3.1 111	W	CEDAR CITY HURRICANE FAULT
1921	SEP 12	10 18			3.7 IV	W	RICHFIELD . TUSHAR FAULT
1921	SEP 12	10 22			3.7 IV	W	RICHFIELD TUSHAR FAULT
1921	SEP 12				3.7 14	W	RICHFIELD TUSHAR FAULT
1921	SEP 13	09 30			4.3 V	W	RICHFIELD TUSHAR FAULT 2 SHOCKS
1921	SEP 28				2.0 11	W	ELSINGRE TUSHAR FLT MANY SHOCKS
	55P 29	14 12				W	ELSINORE TUSHAR FAULT DAMAGE
1921	SEP 30				0.1	W	ELSINORE SEVIER VALLEY
1921						ŵ	ELSINORE
1921	SEP 30	13 30				W	ELSINORE TUSHAR FAULT
1921	OCT 01	02 00				Ŵ	ELSINOPE TUSHAR FAULT DAMAGE
1921	OCT 01	15 32				w	ELSINORE TUSHAR FAULT
1921	OCT 01	17 30				w	ELSINORE TUSHAR FAULT
1921	OCT 02	01 05			5°0 II	W	
1921	OCT 04	03 00			2.0 II	W	ELSINORE TUSHAR FAULT
1921	OCT 04	17 25			2.0 II		ELSINORE TUSHAR FAULT
1921	OCT 0a	20 30			2.0 II	W	
1921	OCT 10	08 30			2.0 I	W	
1921	OCT 10	09 45			2.0 I	W	ELSINORE TUSHAR FAULT
1921	OCT 15				3.1 III	W	ELSINORE TUSHAR FAULT
1921	OCT 15				3.1 III	W	ELSINORE TUSHAR FAULT
1921	OCT 27				5.5 VII	W	ELSINORE TUSHAR FAULT 2 SHOCKS
1921	DEC 20				3.7 IV	W	ELSINORE TUSHAR FAULT
1921	DEC 20				3.7 IV	W	ELSINORE TUSHAR FAULT
1921	DEC 20				3.7 IV	W	ELSINORE TUSHAR FAULT
1921	DEC 20				3.7 IV	W	ELSINORE TUSHAR FAULT
1921	DEC 20				3.7 IV	W	ELSINORE TUSHAR FAULT
1921	DEC 20				3.7 IV	W	ELSINORE TUSHAR FAULT
1923	MAY 14				4.3 V	W	NADA
1923	JUN 07				4.3 V	W	LOGAN EAST CACHE FAULT DAMAGE
1923	JUN 09					W	RICHMOND EAST CACHE FAULT
1923	SEP 07				3.7 IV	w	RICHMOND EAST CACHE FAULT
1924	JAN 01				3.1 III	W	KANE CO ORDERVILLE SEVIER FAULT
1925	JUL 14	13 47			3.7 IV	W	MODENA
1925	DEC 01	07 30			3.1 III	W	SALT LAKE CITY
1925	MAY 03				2.0 II	W	SALT LAKE VALLEY WASATCH FAULT
100	MAY 15	10 51			3.1 III	w	KANE CO ORDERVILLE SEVIER FAULT
1925	JUN 01	05 20			2.0 II	ŵ	ORDERVILLE SEVIER FAULT
1925	JUN 05	11 00			3.1 III	ŵ	ORDERVILLE SEVIER FAULT
1926	JUN 22				2.0 II	w	ORDERVILLE SEVIER FAULT
1925	JUN 28					¥	ORDERVILLE SEVIER FAULT
1925	JUL 12				3.1 III	w	ORDERVILLE SEVIER FAULT
1925	JUL 15					Ŵ	ORDERVILLE SEVIER FAULT
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1925	JUL 29						
1925	OCT 01					~	IFWISTON FAST CACHE FAULT
					3.1 III	¥	LEWISTON EAST CACHE FAULT
		15 15			3.1 III 3.1 III	W	ORDERVILLE SEVIER FAULT
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1925 1925 1925 1927 1927 1927 1928 1931 1931 1932 1932 1932 1932 1932 1933 1934 1934 1934	CCT 04 OCT 23 NOV 12 DEC 19 MAY 22 NOV 22 NOV 23 JUN 02 4AR 09 APR 10 JUN 03 MAY 22 MAR 09 APR 10 JUN 03 MAY 11 DEC 21 JAN 30 MAR 12 MAR 17 APR 17	$\begin{array}{cccccccccccccccccccccccccccccccccccc$		41.7 112.6	3.1 III 3.1 III 2.0 II 2.0 II 2.0 II 2.0 II 2.0 II 2.0 II 2.0 II 2.0 II 2.0 II 3.1 III 3.7 IV 3.1 III 3.1 III 3.7 IV 3.1 III 3.1 III	******************	ORDERVILLE SEVIER FAULT ORDERVILLE SEVIER FAULT ORDERVILLE SEVIER FAULT ORDERVILLE SEVIER FAULT ELHERTA RICHFIELD TUSHAR FAULT ORDERVILLE SEVIER FAULT ORDERVILLE SEVIER FAULT CEDAR CITY USHAR FAULT CEDAR CITY HURRICANE FAULT UND CEDAR CITY HURRICANE FAULT DARDAN SALT LAKE CITY MAGATCH FAULT SALT LAKE CITY
1925 1925 1925 1925 1927 1927 1927 1928 1931 1931 1932 1932 1932 1932 1934 1934 1934	OCT 04 0CT 23 NOV 12 DEC 19 MAY 22 NOV 23 NOV 23 JUN 02 4AY 22 JUN 02 4AY 10 JUN 02 4AY 10 JUN 02 4AY 10 JUN 02 4AY 10 JUN 02 JAN 30 JAN 30 JA	$\begin{array}{cccccccccccccccccccccccccccccccccccc$		41.7 112.6	3.1 III 3.1 III 2.0 II 2.0 II 2.0 II 2.0 II 2.0 II 2.0 II 3.1 III 4.9 VI 3.1 III 3.1 III 3	********************	ORDERVILLE SEVIER FAULT ORDERVILLE SEVIER FAULT ORDERVILLE SEVIER FAULT CUIERTA ELIENTA RICHFIELD TUSHAR FAULT ORDERVILLE SEVIER FAULT ORDERVILLE SEVIER FAULT ELSINGDE TUSHAR FAULT ELSINGDE TUSHAR FAULT CUID ONDERVILLE SEVIER FAULT VENICE PARONAN HURRICANE FAULT VENICE PARONAN HURRICANE FAULT SALT LAKE CITY MARTCH FAULT SALT LAKE CITY MARTCH FAULT SALT LAKE CITY MARTCH FAULT NOSMO AMAGEL VALLEY FAULT NOSMO NAN UTAN HANSEL VALLEY FAULT COLLINISTON WASATCH FAULT

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1934	JUN 02	12.00			2.0	II	w	SALT LAKE CITY WASATCH FAULT
1934	JUL 04				3.7	IV	w	SNOWVILLE MANSEL VALLEY FAULT
					3.7	IV	W	KANAB SEVIER FAULT
1934		10 50			2.0	11	W	NEWTON DAYTON FAULT
1935	MAY 30					II	ŵ	SALT LAKE CITY WASATCH FAULT
1935	JUN 94				2.0			SALT LAKE VALLEY WASATCH FAULT
1935	JUL 09				3.7	IV	W	SALT LARE VALLET WASHICH FAULT
1935	JUL 09				3.7	IV	W	SALT LAKE VALLEY WASATCH FAULT
1935	OCT 05				3.7	IV	×	BOULDER
1935	NOV 06	08 12			2.0	II	W	SALT LAKE CITY WASATCH FAULT
1935	DEC 05	21 15			2.0	II	W	TROPIC PAUNSAGUNT FAULT
1936	MAY 09	10 25			4.3	V	W	ZION NATIONAL PARK 2 SHOCKS
1935	SEP 02				3.1	III	W	KIMBERLY
1937	FEB 13				3.7	TV	W	PANGUITCH, PAROWAN, AND CEDAR CITY
1937	FE8 18				3.7	IV	W	PANGUITCH, PAROWAN, AND CEDAR CITY
1937	FE3 18				3.7	IV	Ŵ	PANGUITCH SEVIER FAULT
					3.1	III	W	PANGUITCH SEVIER FAULT
1937	FEB 21				3.7	IV	ŵ	PANGUITCH SEVIER FAULT
1937	FEB 26							PANGUITCH SEVIER FAULT
1937	MAR 13				2.0	II	W	PANGUITCH SEVIER FAULT
1937	APR 01				2.0	II	W	
1937	NOV 18				3.7	IV	W	LUCIN
1938	MAR 18				3.1	III	w	THISTLE
1938	JUN 30	13 37			4.3	V	w	SALT LAKE VALLEY WASATCH FAULT
1935	JUL 01	18 14			2.0	II	W	SALT LAKE CITY WASATCH FAULT
1938	DEC 03	22 00			2.0	II	W	SALT LAKE CITY WASATCH FAULT
1939	MAR 31				3.7	IV	W	SALT LAKE CITY WASATCH FAULT
1940	FE3 29				2.0	II	W	LOGAN EAST CACHE FAULT
1940	NOV 23)		3.7	IV	Ŵ	MANTI THOUSAND LAKE FAULT
	NOV 23				3.7	IV	Ŵ	MANTI THOUSAND LAKE FAULT
1940		15 22			2.0	ÎÌ	0	MANTI THOUSAND LAKE FAULT
1940	NOV 25	12 30				ÎÌ	2	MANTI THOUSAND LAKE FAULT
1940	NOV 25	13 10			2.0	TV	ŵ	MANTI THOUSAND LAKE FAULT
1940	NOV 25		·		3.7			
1941	JUN 20	15 20			3.1	III	W	LOGAN EAST CACHE FAULT
1942	MAR 28				3.7	IV	w	CIRCLEVILLE TUSHAR FAULT
1942	MAR 28	16 00)		3.7	IV	W	CIRCLEVILLE TUSHAR FAULT
1942	APR 18	06 45			4.3	v	W	HANSEL VALLEY HANSEL VALLEY FAULT
								SANPETE CO EPHRAIM TO NEPHI
1942	JUN 04	23 04			4.3	V	W	
1942	AUG 30	23 08			4.9	VI	W	CEDAR CITY HURRICANE FAULT DAMAGE
1942	SEP 18	01 00			3.1	III	*	CEDAR CITY HURRICANE FAULT
1942	SEP 18	03 00			3.1	III	W	CEDAR CITY HURRICANE FAULT
1942	SEP 18	06 20			3.7	IV	W	CEDAR CITY HUPRICANE FAULT
1942	SEP 26				2.0	IT	W	CEDAR CITY HURRICANE FAULT
1942	SEP 26				4.3	V	W	CEDAR CITY HURRICANE FAULT
1942	SEP 26				2.0	II	W	CEDAR CITY HURRICANE FAULT
1942	SEP 26				4.9	VI	W	CEDAR CITY HURRICANE FAULT DAMAGE
1942	SEP 28				3.7	IV	w	ZION NATIONAL PARK
1942					3.7	IV	w -	CEDAR CITY HURRICANE FAULT DAMAGE
						VI	4	SALT LAKE VALLEY WASATCH FAULT
1943	FEB 22				4.9			HUNTER, MAGNA, AND GARFIELD
1943	FEB 23				2.0	II	W	HUNIER, HAGNA, AND GARFIELD
1943	MAR 12				3.7	IV	W	EPHRAIM THOUSAND LAKE FAULT
1943	APR 10				4.3	V	W	SALT LAKE VALLEY WASATCH FAULT
1943	APR 11	20 32			3.1	III	W	SALT LAKE VALLEY WASATCH FAULT
1943	NOV 03	10 30			4.3	v	W	SEVIER TUSHAR FAULT
1943	NOV 03	12 30			3.1	III	W	SEVIER TUSHAR FAULT
1943	NOV 04				3.1	III	W	SEVIER TUSHAR FAULT
1943	DEC 09				3.7	IV	W	CEDAR CITY HURRICANE FAULT
1943	DEC 09	10 21			3.7	IV	W	CEDAR CITY HURRICANE FAULT
1943	DEC 10				3.7	IV	W	CEDAR CITY HURRICANE FAULT
1943	DEC 10				2.0	II	W	CEDAR CITY HURRICANE FAULT
					3.7	TV	w.	ST GEORGE
1944	MAY 04				2.0	II	ŵ	MONROE SEVIER FAULT
1944	JUN 05							CEDAR CITY HURRICANE FAULT
1944	JUN 13	11 43			2.0	II	W	
1945	MAR 23				2.0	II	W	NEPHI WASATCH FAULT
1945	NOV 18				4.9	VI	W	GLENWOOD SEVIER FAULT DAMAGE
1946	FEB 19	21 15				II	w	PARK VALLEY HANSEL VALLEY FAULT
1946	FEB 20				2.0	II	W	PARK VALLEY HANSEL VALLEY FAULT
1946	MAY 05				4.3	v	w	BEAR RIVER VALLEY AND LOGAN
1940	OCT 25				3.1	III	W	MAGNA
1947	MAR 07				3.1	III	W	SALT LAKE CITY WASATCH FAULT
					3.7	IV	Ŵ	MURRAY WASATCH FAULT
1947	MAR 28	11 02			4.3	1V V	w	MANTI THOUSAND LAKE FAULT
1948	NOV 04	13 18			4.3	v	w	MANTI THOUSAND LAKE FAULT

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1949	MAR 17	06 50		4.9	VI	*	SALT LAKE CITY WASATCH FAULT
1949	MAR 07	06 54			VI	4	SALT LAKE CITY WASATCH FAULT
1949	MAR 07	07 09			IV	W	SALT LAKE CITY WASATCH FAULT
1949	MAR 07	08 06			IV	W	SALT LAKE CITY WASATCH FAULT
1949	MAR 07	03 16			IV	w	SALT LAKE CITY WASATCH FAULT
1949	NOV 22	02 30			VT.	w	WASHINGTON CO HURRICANE FAULT
1950	JAN 02	19 53 04.0	41.5 112.0	3.7	IV	C	NORTHERN UTAH
			40.5 110.5	5.5	1.4	c	NORTHERN UTAH
1950		01 55 31.0		5.5		c	PAYSON
1950	FEB 19					c	PAYSON
1950	FEB 20	14 59		3.7	IV		CENTRAL UTAH SOUTH OF SALT LAKE
1950	FE3 25	13 37 37.0	40.0 112.0			C	
195u	MAY 05	07 35		3.7	IV	C	PIUTE COUNTY
1950	MAY 03	19 40				C	EUREKA
1950	MAY N3	22 35		4.3	V	с	PAYSON DAWAGE
1950	JUL 21	19 23		3.7	IV	С	LOGAN
1951	JAN 23	13 29		3.7	IV	С	NEPHI
1951	JAN 23	13 33		3.7	I٧	с	NEPHI
1951	MAR 05	23 00		3.7	14	с	FREDONIA ARIZONA AND KANAR UTAH
1951	AUG 12	co 25		4.3	v	с	PROVO
1952	MAY 02	10 15		3.1	IIT	с	TOQUERVILLE
1952	MAY 02	15 15		3.1	III	c	TOQUERVILLE
1952	JUL 21	01 00		3.7	IV	c	SANTAGUIN
1952	JUL 23	19 28		3.7	TV	C	SALT LAKE CITY
1952	SEP 28	20 00		4.3	v	c	LEHT
1953	APR 16	05 15		3.7	IV	č	MONROE
1953	MAY 24	02 54 29.0	40.5 111.5	4.3	v	c	NORTH-CENTRAL UTAH SALT LAKE CITY
1953	JUL 30	05 45	40.5 111.5	4.3	v	c	GREENRIVER
1953		15 37		3.7	IV	c	SALT LAKE CITY ROSE PARK AREA
	AUG 16			3.7	IV	č	SALT LAKE CITY ROSE PARK AREA
1953	AUG 16	16 00			IV	c	SALT LAKE CITY ROSE PARK AREA
1953	AUG 16	16 36		3.7		c	PANGUITCH DAMAGE
1953	OCT 22	03 00		4.3	v		
1953	OCT 24	20 50				C	PANGUITCH
1954	MAR 09	11 58		2.0	II	C	SALT LAKE CITY
1954	MAR 31	14 00		3.7	IV	С	GREENRIVER
1954	MAY 12					С	SALT LAKE CITY
1954	MAY 13					с	SALT LAKE CITY
						č	LOGAN AND VICINITY
1954	NOV 01	07 45		4.3	v	č	SALT LAKE CITY REGION DAMAGE
1955	FEB 02	19 23					SOUTH CENTRAL UTAH FRUITA AND TORREY
1955	MAR 27	12 13		3.7	IV	c	CENTERVILLE DAMAGE
1955	MAY 12			4.3			
1955		22 57					
	JUN 25	05 00		3.7	IV	С	MORGAN
1955	NOV 20	05 00	37,1 113,9	3.7	IV	c	SOUTHWEST UTAH
1955 1956	NOV 20 FE3 12	05 00 10 57 00.9 03 00	37.1 113.9	3.7	IV IV	CCC	SOUTHWEST UTAH VERNAL
1955 1956 1956	NOV 20 FEB 12 FEB 12	05 00 10 57 00.9 03 00 04 15	37.1 113.9	3.7 3.7 3.7	IV IV IV	0000	SOUTHWEST UTAH VERNAL VERNAL
1955 1956 1956 1957	NOV 20 FEB 12 FEB 12 JUL 15	05 00 10 57 00.9 03 00 04 15 15 24 20.0	37.1 113.9	3.7	IV IV	00000	SOUTHWEST UTAH VERNAL CENTRAL UTAH
1955 1956 1956 1957 1957	NOV 20 FEB 12 FEB 12 JUL 18 OCT 25	05 00 10 57 00.9 03 00 04 15 15 24 20.0 16 25 47.0	37.1 113.9 40.0 110.5 40.0 111.0	3.7 3.7 3.7	IV IV IV	000000	SOUTHWEST UTAH VERNAL VERNAL CENTRAL UTAH CENTRAL UTAH
1955 1956 1956 1957 1957 1957	NOV 20 FEB 12 FEB 12 JUL 18 OCT 25 OCT 25	05 00 10 57 00.9 03 00 04 15 15 24 20.0 16 25 47.0 01 45 41.0	37.1 113.9 40.0 110.5 40.0 111.0 40.0 111.0	3.7 3.7 3.7 3.7	IV IV IV	0000000	SOUTHWEST UTAH VERNAL CENTRAL UTAH CENTRAL UTAH CENTRAL UTAH
1955 1956 1956 1957 1957 1957 1958	NOV 20 FEB 12 FEB 12 JUL 18 OCT 25 OCT 25 JAN 05	05 00 10 57 00.9 03 00 04 15 15 24 20.0 16 25 47.0 01 45 41.0 17 00 07.4	37.1 113.9 40.0 110.5 40.0 111.0 40.0 111.0 40.0 111.0 40.72 112.4	3.7 3.7 3.7 3.7	IV IV IV IV	200000000000000000000000000000000000000	SOUTHWEST UTAH VERNAL CENTRAL UTAH CENTRAL UTAH CENTRAL UTAH TOOELE CO, 15 MI NW OF TODELE
1955 1956 1956 1957 1957 1957 1958 1958	NOV 20 FEB 12 FEB 12 JUL 18 OCT 25 OCT 25 JAN 05 FEB 13	$\begin{array}{cccccccccccccccccccccccccccccccccccc$	37.1 113.9 40.0 110.5 40.0 111.0 40.0 111.0 40.72 112.4 40.5 111.5	3.7 3.7 3.7 3.7 3.7	IV IV IV	2 2 0 0 0 0 0 0 0 0 0 0 0 0 0 0 0 0 0 0	SOUTHWEST UTAH VERNAL CENTRAL UTAH CENTRAL UTAH CENTRAL UTAH TODELE CO, 15 MI NN OF TODELE 10 MILES NN OF MALLSBURG DAMAGE
1955 1956 1956 1957 1957 1957 1958	NOV 20 FEB 12 FEB 12 JUL 18 OCT 25 JAN 05 FEB 13 VAR 22	05 00 10 57 00.9 03 00 04 15 15 24 20.0 16 25 47.0 01 45 41.0 17 00 07.4 22 52 00.0 15 47 31.8	37.1 113.9 40.0 110.5 40.0 111.0 40.0 111.0 40.0 111.0 40.72 112.4 40.5 111.5 39.72 110.3	3.7 3.7 3.7 3.7 3.7	VI VI VI	20 U D 20 U	SOUTHWEST UTAH VERNAL CENTRAL UTAH CENTRAL UTAH CENTRAL UTAH TOOELE CO. IS MI NW OF TOOELE IO MILES NW OF MALLSBURG DAMAGE CARBON CO. 9 MI N OF SUNNYSIDE
1955 1956 1956 1957 1957 1957 1958 1958	NOV 20 FEB 12 FEB 12 JUL 18 OCT 25 OCT 25 JAN 05 FEB 13	$\begin{array}{cccccccccccccccccccccccccccccccccccc$	37.1 113.9 40.0 110.5 40.0 111.0 40.0 111.0 40.72 112.4 40.5 111.5 39.72 110.3	3.7 3.7 3.7 3.7 3.7 4.9	IV IV IV IV	20 C C C C C C C C C C C C C C C C C C C	SOUTHWEST UTAH VERNAL CENTRAL UTAH CENTRAL UTAH CENTRAL UTAH TOOELE CO. 15 MI NN OF TOOELE 10 MILES NW OF MALLSBURG DAMAGE CARBON CO. 9 MI N OF SUNNYSIDE NEPHI
1955 1956 1956 1957 1957 1957 1958 1958 1958	NOV 20 FEB 12 FEB 12 JUL 18 OCT 25 OCT 25 JAN 05 FEB 13 WAR 22	05 00 10 57 00.9 03 00 04 15 15 24 20.0 16 25 47.0 01 45 41.0 17 00 07.4 22 52 00.0 15 47 31.8	37,1 113,9 40,0 110,5 40,0 111,0 40,0 111,0 40,72 112,4 40,5 111,5 39,72 110,3 40,28 112,4	3.7 3.7 3.7 3.7 3.7 4.9 4.3 4.3	IV IV IV VI V	50 00 00 00 00 00 00 00 00 00 00 00 00 0	SOUTHWEST UTAH VERNAL CENTRAL UTAH CENTRAL UTAH TOOELE CO. 15 MI NW OF TODELE 10 MILES NW OF WALLSBURG DAMAGE CARBON CO. 9 MI NO F SUNNYSIDE NEPHI TOOELE CO. 51. JONN AREA DAMAGE
1955 1956 1956 1957 1957 1957 1958 1958 1958 1958 1955	NOV 20 FEB 12 FEB 12 JUL 18 OCT 25 JAN 05 FEB 13 VAR 22 NOV 28	$\begin{array}{cccccccccccccccccccccccccccccccccccc$	37,1 113,9 40,0 110,5 40,0 111,0 40,0 111,0 40,72 112,4 40,5 111,5 39,72 110,3 40,28 112,4	3.7 3.7 3.7 3.7 3.7 3.7 3.7 3.7 3.7 3.7	IV IV IV VI V	20 20 20 20 20 20 20 20 20 20 20 20 20 2	SOUTHWEST UTAH VERNAL VERNAL ENTAL UTAH CENTRAL UTAH CENTRAL UTAH TOOELE CO. IS MI NW OF TOOELE IO MILES NW OF WALLSBURG DAWAGE CARBON CO. IS MI NW OF TOOELE CO. ST. JOHN AREA DAWAGE TOOELE CO. ST. JOHN AREA DAWAGE
1955 1956 1957 1957 1957 1958 1958 1958 1958 1958 1955 1955	NOV 20 FEB 12 FEB 12 JUL 18 OCT 25 JAN 05 FEB 13 VAR 22 NOV 29 DEC 01	$\begin{array}{cccccccccccccccccccccccccccccccccccc$	37.1 113.9 40.0 110.5 40.0 111.0 40.0 111.0 40.72 112.4 40.5 111.5 39.72 110.3	3.7 3.7 3.7 3.7 3.7 3.7 3.7 3.7 3.7 3.7	IV IV IV VI V	20 20 20 20 20 20 20 20 20 20 20 20 20 2	SOUTHWEST UTAH VERNAL CENTRAL UTAH CENTRAL UTAH TOOELE CO. 15 MI NW OF TODELE 10 MILES NW OF WALLSBURG DAMAGE CARBON CO. 9 MI NO F SUNNYSIDE NEPHI TOOELE CO. 51. JONN AREA DAMAGE
1955 1956 1957 1957 1957 1958 1958 1958 1958 1955 1955 1955	NOV 20 FEB 12 FEB 12 JUL 18 OCT 25 JCT 25 JCT 25 JCT 25 FEB 13 VAR 22 NOV 28 DEC 01 DEC 01 DEC 01	$\begin{array}{cccccccccccccccccccccccccccccccccccc$	37.1 113.9 40.0 110.5 40.0 111.0 40.0 111.0 40.7 112.4 40.5 112.4 40.5 112.5 40.28 112.5	3.7 3.7 3.7 3.7 3.7 3.7 3.7 3.7 3.7 3.7	VI VI VI VI VI VI VI	00000000000000000000000000000000000000	SOUTHWEST UTAH VERNAL VERNAL ENTAL UTAH CENTRAL UTAH CENTRAL UTAH TOOELE CO. IS MI NW OF TOOELE IO MILES NW OF WALLSBURG DAWAGE CARBON CO. IS MI NW OF TOOELE CO. ST. JOHN AREA DAWAGE TOOELE CO. ST. JOHN AREA DAWAGE
1955 1956 1957 1957 1957 1958 1958 1958 1958 1955 1955 1955	NOV 20 FEB 12 JUL 15 OCT 25 OCT 25 OCT 25 FEB 13 VAR 22 DEC 01 DEC 01 DEC 02	$\begin{array}{cccccccccccccccccccccccccccccccccccc$	37,1 113.9 40.0 110.5 40.0 110.5 40.0 111.0 40.0 111.0 40.0 111.0 39.72 112.4 40.5 111.5 39.72 110.3 40.45 112.5 40.45 112.5 40.45 112.5 40.45 112.5	3.7 3.7 3.7 3.7 4.9 4.3 4.3 5 3.7 4.3	VI VI VI VV VV VV VV VV	00000000000000000000000000000000000000	SOUTHWEST UTAH VERNAL CENTRAL UTAH CENTRAL UTAH CENTRAL UTAH CENTRAL UTAS IN HILE NA OF TOOELE IN WILE NA OF NUMBYSIDE CARBON CO. 9 MIN OF STUNNYSIDE NEPHI TOOELE CO. 5 JOHN AREA DAMAGE TOOELE CO. 6 MIN NO F ST. JOHN
1955 1956 1957 1957 1957 1958 1958 1958 1958 1958 1958 1958	NOV 20 FEB 12 FEB 12 JUL 15 OCT 25 OCT 25 OCT 25 FEB 13 VAR 22 NOV 28 DEC 01 DEC 01 DEC 01 DEC 02 DEC 02 DEC 11	$\begin{array}{cccccccccccccccccccccccccccccccccccc$	37,1 113,9 40,0 110,5 40,0 111,0 40,0 111,0 40,0 111,0 40,0 111,0 40,0 111,0 40,0 111,0 39,72 110,3 40,28 112,4 40,45 112,5 40,45 112,5 40,45 112,5 40,45 112,5	3.7 3.7 3.7 3.7 3.7 3.7 3.7 3.7 3.7 3.7	IV IV IV V IV IV IV IV IV	00000000000000000000000000000000000000	SOUTHWEST UTAH VERNAL VERNAL CENTRAL UTAM CENTRAL UTAM TODELE CO. 15 MI NW OF TOOFLE 10 MILES NW OF WALLSBURG DAMAGE CARBON CO. 9 MI N OF SUMMYSIDE MOOFLE CO. 5T. JOHN AREA DAMAGE TOOFLE CO. 6 MI NW OF ST. JOHN TOOFLE CO. 6 MI NW OF ST. JOHN TOOFLE CO. 8 MI W OF ST. JOHN
1955 1956 1957 1957 1957 1958 1958 1958 1958 1958 1958 1958 1958	NOV 20 FEB 12 JUL 18 OCT 25 JUL 18 OCT 25 JAN 05 FEB 13 NOV 28 DEC 01 DEC 01 DEC 01 DEC 01 DEC 11 DEC 14	$\begin{array}{cccccccccccccccccccccccccccccccccccc$	37.1 113.9 40.0 110.5 40.0 111.0 40.72 112.4 40.5 111.5 39.72 110.3 40.28 112.4 40.45 111.5 39.72 110.3 40.45 112.4 40.45 112.5 40.46 112.6 40.45 112.5 40.45 112.6 40.45 112.6	3.7 3.7 3.7 3.7 3.7 4.9 4.3 3.4.3 5.3.7 4.3 3.7 3.7 4.3 3.7	VI VI VI VV VV VV VV VV	5000000 5000000 50000000 500000000 5000000	SOUTHWEST UTAH VERNAL VERNAL ENTRAL UTAH CENTRAL UTAH CENTRAL UTAH TOOELE CO. 15 MI NW OF TOOELE 10 MILES NW OF MALLSBURG DAMAGE CARGON CO. 15 MI NW OF STOUHANDE NEPHI TOOELE CO. 6 MI NW OF ST. JOHN TOOELE CO. 6 MI NW OF ST. JOHN
1955 1956 1957 1957 1957 1958 1958 1958 1958 1958 1958 1958 1958	NOV 20 FEB 12 JUL 18 JUL 18 JUCT 25 JUCT 25 JAN 05 FEB 12 JAN 04 DEC 01 DEC 01 DEC 02 DEC 114 DAN 04	$\begin{array}{cccccccccccccccccccccccccccccccccccc$	37.1 113.9 40.0 110.5 40.0 111.0 40.7 11.0 40.7 11.0 40.7 11.5 39.72 110.3 40.28 112.4 40.45 112.5 40.45 112.5 40.46 112.5	3.7 3.7 3.7 3.7 4.9 4.3 4.3 5 3.7 4.3	IV IV IV IV IV IV IV IV IV IV		SOUTHWEST UTAH VERNAL VERNAL CENTRAL UTAH CENTRAL UTAH TOOELE CO. IS MI NW OF TOOELE ID MILES NW OF MALLSBURG DAMAGE CARBON CO. IS MI NW OF STUDYSTOE NEPHI TOOELE CO. 5 MI NW OF ST. JOHN TOOELE CO. 5 MI NW OF ST. JOHN MADIE CO. 8 MI W OF ST. JOHN WATTIS CO. 8 MI W OF ST. JOHN WATTIS CO. 8 MI W OF ST. JOHN
1955 1956 1957 1957 1957 1957 1958 1958 1958 1958 1958 1958 1958 1958	NOV 20 FEB 12 JUL 18 JUL 18 OCT 25 JAN 05 FEB 13 JAN 05 FEB 13 DEC 01 DEC 01 DEC 02 DEC 14 JAN 04 FEB 27	$\begin{array}{cccccccccccccccccccccccccccccccccccc$	37.1 113.9 40.0 110.5 40.0 11.0 40.0 11.0 40.72 112.4 40.5 11.5 30.72 112.4 40.45 11.5 40.45 112.5 40.45 112.5 40.45 112.5 40.45 112.5 50.0 112.6	3.7 3.7 3.7 3.7 3.7 4.9 4.3 3.7 4.3 3.7 4.3 3.7 4.3 3.7 4.9	V IV IV IV V IV V V V V V V V V V V V V	50 0 0 0 0 0 0 0 0 0 0 0 0 0 0 0 0 0 0	SOUTHWEST UTAH VERNAL VERNAL CENTRAL UTAM CENTRAL UTAM CENTRAL UTAM TOOELE CO. 15 MI NW OF TOOELE IO MILES NW OF MALLSBURG DAMAGE CARBON CO. 9 MI N OF SUNWSIDE MECHI TOOELE CO. 5 MINN OF ST. JOHN TOOELE CO. 6 MI NW OF ST. JOHN TOOELE CO. 8 MI W OF ST. JOHN TOOELE CO. 8 MI W OF ST. JOHN TOOELE CO. 8 MI W OF ST. JOHN
1955 1956 1957 1957 1957 1958 1958 1958 1958 1958 1958 1958 1958	NOV 20 FEB 12 JUL 18 JUL 18 JUCT 25 JUCT 25 JU	$\begin{array}{cccccccccccccccccccccccccccccccccccc$	37.1 113.9 40.0 110.5 40.0 111.0 40.0 111.0 40.72 112.4 40.73 112.4 40.8 112.4 40.45 112.4 40.46 112.6 39.72 110.3 39.72 110.3 39.72 112.4 40.45 112.5 40.46 112.6 35.0 112.5 37.0 112.5	3.7 3.7 3.7 3.7 3.7 3.7 3.7 4.9 4.3 3.7 3.7 4.3 3.7 4.3 3.7 4.3 5.7 4.3 5.7 4.3 5.7 4.3 5.7 4.5 5.7 5.7 5.7 5.7 5.7 5.7 5.7 5	IV IV IV VI VV VV VV VV VV VV VV VV VV	0 0 0 0 0 0 0 0 0 0 0 0 0 0 0 0 0 0 0	SOUTHWEST UTAH VERNAL VERNAL VERNAL CENTRAL UTAH CENTRAL UTAH CENTRAL UTAH TOOELE CO. 15 MI NW OF TOOELE 10 MILES NW OF MALLSBURG DAMAGE CARGON CO. 15 MI NW OF STOUHH TOOELE CO. 6 MI NW OF ST. JOHN TOOELE CO. 6 MI NW OF ST. JOHN NETHI TOOELE CO. 6 MI NW OF ST. JOHN WATTIS NETHI SE UTAH PANGUITCH DAMAGE SE UTAH PANGUITCH DAMAGE
1955 1956 1957 1957 1957 1957 1958 1958 1958 1958 1958 1958 1958 1958	NOV 20 FEB 12 JUL 15 OCT 25 OCT 25 OCT 25 OCT 25 FEB 13 VAR 22 DEC 01 DEC 01 DEC 01 DEC 01 DEC 01 DEC 01 DEC 11 DEC 11 DEC 11 DEC 11 DEC 27 JUL 27	$\begin{array}{cccccccccccccccccccccccccccccccccccc$	37.1 113.9 40.0 110.5 40.0 11.0 40.0 11.0 40.72 112.4 40.5 11.5 30.72 112.4 40.45 11.5 40.45 112.5 40.45 112.5 40.45 112.5 40.45 112.5 50.0 112.6	3.7 3.7 3.7 3.7 3.7 3.7 4.9 4.3 5.4.9 4.3 5.7 4.3 5.7 4.9 5.7 4.9 5.7 4.9 5.7 4.9 5.7 4.9 5.7 4.9 5.7 5.7 5.7 5.7 5.7 5.7 5.7 5.7	IV IV IV IV IV IV IV IV IV IV IV IV IV		SOUTHWEST UTAH VERNAL CENTRAL UTAH CENTRAL UTAH TOOELE CO. 15 MI NW OF TODELE IO MILES NW OF WALLSBURG DAMAGE CARBON CO. 15 MI NW OF TODELE IO MILES NW OF WALLSBURG DAMAGE TODELE CO. 5 N. JOHN AREA DAMAGE TODELE CO. 6 MI NW OF ST. JOHN TODELE CO. 6 MI NW OF ST. JOHN TODELE CO. 6 MI NW OF ST. JOHN TODELE CO. 8 MI W OF ST. JOHN MATTIS DIAHO N. OF DEAR LAWE DAMAGE UTAH-ARIZIONA BORDER KANAB DAMAGE
1955 1956 1957 1957 1957 1958 1958 1958 1958 1958 1958 1958 1958	NOV 20 FEB 12 JUL 15 OCT 25 OCT 25 OCT 25 FEB 12 JAN 05 FEB 12 FEB 12 FEB 12 SCC 25 OCT 25 OC	$\begin{array}{cccccccccccccccccccccccccccccccccccc$	37.1 113.9 40.0 110.5 40.0 111.0 40.0 111.0 40.72 112.4 40.55 111.5 39.72 110.3 40.85 112.4 40.45 112.4 40.45 112.5 40.45 112.5 40.46 112.6 35.0 112.5 37.0 112.5 37.0 112.5	3.77 3.77 3.77 3.77 3.77 3.77 4.99 4.33 4.35 5.77 4.39 5.77 3.77 4.39 5.77 3.77 4.39 5.77 3.77 4.39 5.777 5.7777 5.7777 5.7777 5.7777 5.7777 5.7777 5.77777 5.7777 5.7777 5	IV IV IV VI VV VV VV VV VV VV VV VV VV		SOUTHWEST UTAH VERNAL VERNAL CENTRAL UTAM CENTRAL UTAM CENTRAL UTAM TOOELE CO. 15 MI NW OF TOOELE 10 MILES NW OF MALLSBURG DAMAGE CARBON CO. 15 MI NW OF ST. JOHN NEPHI TOOELE CO. 6 MI NW OF ST. JOHN TOOELE CO. 6 MI NW OF ST. JOHN NEPHI SE 10AHO N OF BEAR LAYE DAMAGE SE UTAH PANGUITCH DAMAGE UTAMARAITONE GONDER KANAB DAMAGE UTAMARAITONE GONDER KANAB DAMAGE
1955 1956 1957 1957 1957 1958 1958 1958 1958 1958 1958 1958 1958	NOV 20 FE3 12 FE3 12 JUL 15 OCT 25 JUL 15 OCT 25 FE3 13 V4R 22 NOV 25 DEC 01 DEC 01 DEC 01 DEC 01 DEC 14 JUL 27 SEP 17 MAY 06	$\begin{array}{cccccccccccccccccccccccccccccccccccc$	37.1 113.9 40.0 110.5 40.0 110.5 40.0 110.7 40.72 112.4 40.65 111.5 30.72 110.3 40.46.5 112.5 40.45 112.5 37.0 112.5 30.94 110.4	3.77 3.77 3.77 3.77 3.77 3.77 4.99 4.33 4.35 5.77 4.39 5.77 3.77 4.39 5.77 3.77 4.39 5.77 3.77 4.39 5.777 5.7777 5.7777 5.7777 5.7777 5.7777 5.7777 5.77777 5.7777 5.7777 5	IV IV IV IV IV IV IV IV IV IV IV IV IV		SOUTHWEST UTAH VERNAL CENTRAL UTAH CENTRAL UTAH ETATRAL UTAH ISODELE CO.15 MI NW OF TODELE IO MILES NW OF WALLSBURG DAMAGE CARBON CO.05 MI NW OF STUNNYSIDE NEPHI TODELE CO. 6 MI NW OF ST. JOHN TODELE CO. 6 MI NW OF ST. JOHN TODELE CO.6 MI NW OF ST. JOHN TODELE CO.8 MI W OF ST. JOHN MATTIS EUTAH PANGUITCH DAMAGE SUTAH-ASIZONA BORDER KANAB DAMAGE KANNRARVILE MARYSVALE CARBON CO. 10 MI SE OF GOLDIEP SUMMIT
1955 1956 1957 1957 1958 1958 1958 1958 1958 1958 1958 1958	NOV 20 FEB 12 FEB 12 JUL 15 OCT 25 JUL 15 OCT 25 JUL 15 OCT 25 FEB 13 VMR 22 DEC 01 DEC 01 DEC 01 DEC 01 DEC 01 DEC 01 DEC 11 JAN 04 FEB 22 LOVE 21 JUL 27 SEP 17 AV 06 JUL 27 SEP 17 OCT 25 JUL 21 JUL 27 SEP 17 AV 06 JUL 27	$\begin{array}{cccccccccccccccccccccccccccccccccccc$	37.1 113.9 40.0 110.5 40.0 111.0 40.7 111.0 40.7 111.0 40.7 112.5 40.8 112.5 40.8 112.5 40.46 112.5 40.45 112.5 30.0 112.5 37.0 112.5 37.0 112.5 37.0 112.5 37.0 112.5 37.0 112.5 37.0 112.5 37.0 112.5	3.77 3.77 3.77 3.77 3.77 3.77 4.99 4.33 4.35 5.77 4.39 5.77 3.77 4.39 5.77 3.77 4.39 5.77 3.77 4.39 5.777 5.7777 5.7777 5.7777 5.7777 5.7777 5.7777 5.77777 5.7777 5.7777 5	IV IV IV IV IV IV IV IV IV IV IV IV IV	50 0 0 0 0 0 0 0 0 0 0 0 0 0 0 0 0 0 0	SOUTHWEST UTAH VERNAL VERNAL CENTRAL UTAH CENTRAL UTAH TOOELE CO. IS MI NW OF TOOELE ID MILES NW OF MALLSBURG DAMAGE CARBON CO. IS MI NW OF STOUNHSIDE NEPHI TOOELE CO. G MI NO F ST. UDHH TOOELE CO. G MI NW OF ST. UDHH TOOELE CO. G MI NW OF ST. UDHH MATTIS SE DAHO N OF BEAR LAKE DAMAGE SE UTAH PANGUITCH DAMAGE SE UTAH PANGUITCH DAMAGE UTAH-ABTIZONE BOADER KANAB DAMAGE CARBON CO. ID MI SE OF SOLDIEP SUMMIT NOTHWESTEN UTAH
1955 1956 1956 1957 1957 1958 1958 1958 1958 1958 1958 1958 1958	NOV 20 FE3 12 FE3 12 JUL 15 OCT 25 JUL 15 OCT 25 JAN 05 FE3 13 V4R 22 NOV 25 DEC 01 DEC 01 DEC 01 DEC 01 DEC 01 DEC 01 DEC 01 DEC 14 JUL 27 SEP 12 JUL 27 SEP 10 JUL 09 SEP 12	$\begin{array}{cccccccccccccccccccccccccccccccccccc$	37.1 113.9	3.77 3.77 3.77 3.77 3.77 3.77 4.99 4.33 4.35 5.77 4.39 5.77 3.77 4.39 5.77 3.77 4.39 5.77 3.77 4.39 5.777 5.7777 5.7777 5.7777 5.7777 5.7777 5.7777 5.77777 5.7777 5.7777 5	IV IV IV IV IV IV IV IV IV IV IV IV IV	C C C C C C C C C C C C C C C C C C C	SOUTHWEST UTAH VERNAL CENTRAL UTAH CENTRAL UTAH CENTRAL UTAH CENTRAL UTAH GENTRAL UTAS NEWSITIAL WISH NAY OF TOOELE IO MILES NW OF WALLSBURG DAMAGE CARBON CO. 9 MI NO F TOOELE CO. NEPHI TOOELE CO. 9 MI NO F ST. JOHN TOOELE CO. 6 MI NW OF ST. JOHN MATTIS SIDIAH PANGUITCH DAMAGE SIDIAH PANGUITCH DAMAGE SIDIAH PANGUITCH DAMAGE MANRAPAILLE MARYSALLE CARBON CO. 10 MI SE OF SOLDIEP SUMMIT NOTHWESTERN UTAH
1955 1956 1956 1957 1957 1957 1958 1958 1958 1958 1958 1958 1958 1958	NOV 20 FEB 12 FEB 12 JUL 15 OCT 25 JUL 15 OCT 25 JUL 25 FEB 13 WAR 22 DEC 01 DEC 01 DEC 01 DEC 01 DEC 01 DEC 01 DEC 11 JAN 04 FEB 21 FEB 21 FEB 21 JUL 09 SEP 12	$\begin{array}{cccccccccccccccccccccccccccccccccccc$	37.1 113.9 40.0 110.5 40.0 110.5 40.1 110.0 40.7 112.5 40.8 112.5 40.8 112.5 40.40.5 112.5 40.45 112.5 40.45 112.5 40.45 112.5 57.0 112.5 37.0 112.5 37.0 112.5 37.0 112.5 37.0 112.5 37.0 112.5 37.0 112.5 37.0 112.5 37.0 112.5 37.0 112.5 37.0 112.5 37.0 112.5 37.0 112.5 37.0 112.5 37.0 112.5 37.0 112.5 38.9 111.5	3.7 3.7 3.7 3.7 3.7 3.7 4.3 3.7 4.3 3.7 4.3 5.7 3.7 4.3 5.7 3.7 4.3 5.7 3.7 3.7 4.3 5.7 3.7 3.7 3.7 3.7 3.7 3.7 3.7 3	IV IV IV IV IV IV IV IV IV IV IV IV IV		SOUTHWEST UTAH VERNAL VERNAL CENTRAL UTAH CENTRAL UTAH GENTRAL UTAH TOOELE CO. IS MI NW OF TOOELE ID MILES NW OF MALLSBURG DAMAGE CARBON CO. IS MI NW OF SUNWYSIDE NEPHI TOOELE CO. ST. JOHN AREA DAMAGE TOOELE CO. 5 MI NW OF ST. JOHN WITTOOELE CO. 6 MI NW OF ST. JOHN WITTOOELE CO. 8 MI W OF ST.
1955 1956 1957 1957 1957 1958 1958 1958 1958 1958 1958 1958 1958	NOV 20 FEB 12 FEB 12 JUL 15 OCT 25 JUL 25 FEB 13 VAR 22 FEB 13 VAR 22 NOV 28 DEC 01 DEC 01 DEC 01 DEC 01 DEC 01 DEC 01 DEC 01 L 25 SEP 17 JUL 27 SEP 17 JUL 27 SEP 17 JUL 27 SEP 17 SEP 12 JUL 27 FEB 21 SEP 12 JUL 27 SEP 12 SEP	$\begin{array}{cccccccccccccccccccccccccccccccccccc$	37.1 113.9 40.0 110.5 40.0 110.5 40.0 110.5 40.0 111.0 40.0 111.0 40.0 111.0 40.0 111.0 40.8 112.5 40.45 112.5 40.45 112.5 40.45 112.5 57.0 112.5 37.0 112.5 37.0 112.5 35.9.0 112.4 37.0 112.5 37.9 112.5 37.0 112.5 37.9 112.5 37.9 112.5 37.9 112.5 37.9 112.5 37.9 110.4	3.77 3.77 3.77 3.77 3.77 3.77 4.99 4.33 4.35 4.35 4.35 5.77 3.77 4.99 5.77 3.77 4.99 5.77 3.77 4.99 5.77 3.77 4.99 5.77 3.77 3.77 5.777 5.7777 5.7777 5.7777 5.7777 5.7777 5.7777 5.77777 5.7777 5.77777 5.77777 5.777777 5.7777777777			SOUTHWEST UTAH VENNAL VENNAL VENNAL CENTRAL UTAH CENTRAL UTAH CENTRAL UTAH CENTRAL UTAH TOOELE CO. 15 MI NW OF TOOELE 10 MILES NW OF MALLSBURG DAMAGE CARBON CO. 15 MI NW OF ST. JOHN NEPHI TOOELE CO. 6 MI NW OF ST. JOHN NEPHI SE UTAH PANGUITCH DAMAGE SE UTAH PANGUITCH DAMAGE KANARRAVILLE KARARAVILE CARBON CO. 10 MISE OF SOLDIEP SUMMIT NOTHWESTERN UTAH CENTRAL UTAH CENTRAL UTAH CENTRAL UTAH
1955 1956 1956 1957 1957 1958 1958 1958 1958 1958 1958 1958 1958	NOV 20 FEB 12 JUL 18 CCT 25 OCT 25 FEB 12 JAN 05 FEB 12 VAR 22 NOV 25 CCT 21 DEC 01 DEC 01 DEC 01 DEC 01 DEC 01 DEC 01 DEC 14 JUL 27 SEP 10 CT 27 DEC 01 DEC 14 JUL 27 SEP 10 SEP	$\begin{array}{cccccccccccccccccccccccccccccccccccc$	37.1 113.9 40.0 110.5 40.0 111.0 40.1 110.5 40.0 111.0 40.1 112.5 30.7 112.4 40.4 112.5 40.46 112.5 40.46 112.5 57.0 112.5 50.9 110.45 50.9 110.45 50.9 112.5 50.9 112.5 50.3 112.5 50.3 112.5 50.3 112.5 50.3 112.5 50.3 112.5 50.3 112.5 50.3 112.5 50.5 110.2 50.5 110.2	3.7 3.7 3.7 3.7 3.7 3.7 4.3 3.7 4.3 3.7 4.3 5.7 3.7 4.3 5.7 3.7 4.3 5.7 3.7 3.7 4.3 5.7 3.7 3.7 3.7 3.7 3.7 3.7 3.7 3	IV IV IV IV IV IV IV IV IV IV IV IV IV		SOUTHWEST UTAH VERNAL VERNAL CENTRAL UTAH CENTRAL UTAH TOOELE CO. IS MI NW OF TOOELE ID MILES NW OF MALLSBURG DAMAGE CARBON CO. IS MI NW OF STOUMYSIDE NEPHI TOOELE CO. G MI N OF ST. JOHN TIOELE CO. 6 MI NW OF ST. JOHN TIOELE CO. 6 MI NW OF ST. JOHN TATTIS CO. 8 MI W OF ST. JOHN TATTIS CO. 9 MI N OF ST. JOHN TATTIS CO. 9 MI W OF ST. 9 MIN SCORES CO. 9 MIN SCORES CO. 9 MI W OF ST. 9 MIN SCORES CO. 9
1955 1956 1957 1957 1957 1958 1958 1958 1958 1958 1958 1958 1958	NOV 20 FEB 12 FEB 12 JUL 15 OCT 25 JUL 25 FEB 13 VAR 22 FEB 13 VAR 22 NOV 28 DEC 01 DEC 01 DEC 01 DEC 01 DEC 01 DEC 01 DEC 01 L 25 SEP 17 JUL 27 SEP 17 JUL 27 SEP 17 JUL 27 SEP 17 SEP 12 JUL 27 FEB 21 SEP 12 JUL 27 SEP 12 SEP	$\begin{array}{cccccccccccccccccccccccccccccccccccc$	37.1 113.9 40.0 110.5 40.0 110.5 40.0 110.5 40.0 111.0 40.0 111.0 40.0 111.0 40.0 111.0 40.8 112.5 40.45 112.5 40.45 112.5 40.45 112.5 57.0 112.5 37.0 112.5 37.0 112.5 35.9.0 112.4 37.0 112.5 37.9 112.5 37.0 112.5 37.9 112.5 37.9 112.5 37.9 112.5 37.9 112.5 37.9 110.4	3.77 3.77 3.77 3.77 3.77 4.99 4.33 4.37 5.4.9 5.77 3.77 4.39 5.77 3.77 4.39 5.77 3.77 4.39 5.77 3.77 4.99 5.777 5.7777 5.777 5.7777 5.7777 5.7777 5.7777 5.77777 5.77777 5.777777 5.7777777777		C C C C C C C C C C C C C C C C C C C	SOUTHWEST UTAH VENNAL VENNAL VENNAL CENTRAL UTAH CENTRAL UTAH CENTRAL UTAH CENTRAL UTAH TOOELE CO. 15 MI NW OF TOOELE 10 MILES NW OF MALLSBURG DAMAGE CARBON CO. 15 MI NW OF ST. JOHN NEPHI TOOELE CO. 6 MI NW OF ST. JOHN NEPHI SE UTAH PANGUITCH DAMAGE SE UTAH PANGUITCH DAMAGE KANARRAVILLE KARARAVILE CARBON CO. 10 MISE OF SOLDIEP SUMMIT NOTHWESTERN UTAH CENTRAL UTAH CENTRAL UTAH CENTRAL UTAH

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D REMARKS

YEAR	мо	DAY	HR	MN	SEC	LAT-14	LONG-W	MAG	INT DE		5 D	REMARKS
1961	OCT	15	19	13	06.5	39.2	111.5			e (CENTRAL UTAH
1961	OCT	17	c:n	59	41.8	39.2	111.5		5	1 (9	CENTRAL UTAH
1951	OCT	17	0.3	54	30.3	39.23	111.57					SAN PETE CO. 5 MI SE OF MANTI
1962	FEB	15	07	12	42.9	36.9	112.4	4.5		6 0		UTAH-ARIZONA BORDER UTAH-ARIZONA BORDER SW OF KANAB
1962	FEB	15			45.1	37.0	112.9			1 (
1962	FE3	17	0.3	20	25.9	39.6	111.5			3 (NE OF PANSUITCH
1962	JUN	03	19	15	62.0	38.0	113.0			2.0		NW OF PAROWA"
1962	JUN		00	17	0.0	37,5	113.0		3	3 (<u> </u>	SOUTHERN UTAH
1962	JUN				45.0	39.0	112.1		3	3.5	5	NE OF PANGUITCH
1962	JUN	0.9			07.0	37.5	112.1 113.0 112.5		3	2.5	C .	SE OF CEDAR CITY S OF PANGUITCH
1962	JUN				27.0	37.5	112.5		3	3 5		SW OF SUNNYSIDE
1962	JUN				10.0	30.5	110.5		3			S OF PANGUITCH
1962	JUN		0.0	47	35.0	37.5	112.5		2			UTAH CO, LAKE MTNS.
1962	JUL		67	03	07.0	40.24	111.89			0 1		SALT LAKE CO, 16 MI W OF SLC
1962	JUL				57.0				5	n 1	ii ii	EMERY CO, 6 MI S OF EMERY
1962	AUG				44.6	38.95	111.27	3.4	2	0 1	ii ii	EMERY CO. & MI NW OF EMERY
1952	AUG			05			111.56		2	0 1	1	WAYNE CO, 12 MI S OF LOA
1962	AUG				42.9		111.0		3	3 1	C C	CENTRAL UTAH N OF CASTLE DALE
1962	AUG				28.8	41 22	111.79	5.7	4	0 1	UC	1 MI E OF RICHMOND DAMAGE
1962	AUG		10	55	15.6	41.73	111.96	3.8		0 1		CACHE CO, 1 MI W OF LOGAN
1962	AUG		10	33	32.2	42.13	111.07	3.9	2	0 1	U	S IDAHO, FRANKLIN CO
1962	AUG		18	41	35.7	30.04	112.20		- 2	0 1	U	TODELE CO, 6 MI W OF EUREKA
1962		31			58.2		112.21			0 !		TOOELE CO, 6 MI NW OF EUREKA
1962	SEP				41.7		112.16		2	0 1	U	UTAH CO. 6 MI NE OF EUREKA
1962	SEP	05	16	04	29.1	40.73	112.11	5.2				SALT LAKE CO, MAGNA AREA DAMAGE
1962	SEP				33.7		112.01		1	0	U	SALT LAKE CO, KEARNS AREA
1962	SEP				19.0		111.0			3		SE OF STRAWBERRY RESEVOIR
1962	SEP		16	50	28.0	39.5	110.9			3		D SE OF PRICE
1962	SEP	08			47.6	42.15	112.03			0		S IDAHO, FRANKLIN CO
1962	SEP				43.4	41.09	112.44	1.1	1	0	υ.	WEBER CO, GREAT SALT LAKE D CACHE CO, 1 MI NE OF LOGAN
1962	SEP				12.1	41.76	111.79	4.0	4	0	0.5	JUAB CO, 12 MI SW OF NEPHI
1962	SEP				38.4		112.04	2.0		0		BOX ELDER CO, PROMONTORY POINT AREA
1962	SEP	14	00	29	53.1		112.51		2	0		S IDAHO, 8 MI NE OF RICHMOND, UTAH
1962	SEP	14	13	16	59.1	42.03	111.76	3.9		v .	0	
1962	SEP	30	09	37	28.3	41.29	111.98	2.3		0 1		WEBER CO, 6 MI N OF OGDEN
1962	OCT	14	07	35	52.7		111.10	2.5		0ι		EMERY CO, W OF HIAWATHA
1962	NOV	24	13	14	23.6		111.96			0 (UTAH CO, W OF AMERICAN FORK
1962	NOV				54.7		112,46			0 1		TOOELE CO, 18 MI W OF EUREKA
1962	DEC				37.0	40.2	110.6		3.	3 (-	NORTHEASTERN UTAH
1962	DEC				18.8		110.63	4.3			10	CARBON CO, 12 MI E OF PRICE CARBON CO, 3 MI W OF WELLINGTON
1963	JAN		13	51	29.6		110.78			3 (2	WEST OF CEDAR CITY
1963	FEB		17	34	29.0		112.22	4.0	2		ĭ	DAVIS CO, ANTELOPE ISLAND
1963 1963	MAR		23	40	23.7		110.78			0 1	I D	CARBON CO, S OF PRICE
1963	MAR				37.2		111.90		2	0 1	υŏ	SAN PETE CO, 14 MI W OF MANTI
1963	APR				27.1		111.2		3	3 (Ċ.	SOUTHERN IDAHO
1963	APR		22	18	24.6		110.64	4.2	1	01	JD	CARBON CO, 16 MI E OF HUNTINGTON
1963	APR		13	33	07.1	39.69	110.47	4.6)	0 1	JD	CARBON CO, 12 MI N OF DRAGERTON
1963	MAY	06			00.1		110.0					CARBON CO, PROBABLE ROCKBURST
1963	MAY	24	14	30	20.5		111.72			0 (SALT LAKE CO, 15 MI SE OF SALT LAKE
1963	JUN		08				112.5	4.2		5 (NW OF PANGUITCH
1963	JUN				22.8		112.05	3.0	21	0 1		DAVIS CO, 5 MI SW OF KAYSVILLE
1963	JUL				42.4	39.55	111.92	4.9	3	0 1	U D	JUAB CO, 2 MI W OF LEVAN DAMAGE JUAB CO, 12 MI NW OF NEPHI
1963	JUL				46.9		111.93	3.6	2		0	UTAH CO, S OF STRAWBERRY RESERVOIR
1963	JUL.				26.7		111.23	4 - 1	1		0 0	JUAB CO, 12 MI W OF LEVAN
1963	JUL				34.5		112.16	3.4	2		10	UTAH CO, SW OF STRAWBERRY RESERVOIR
1963	JUL		18		17.1		110.44	3.7		0 1	i n	CARBON CO, 6 MI N OF DRAGERTON
1963 1963	AUG				04.1		112.23	3.7				BOX ELDER CO, NEAR TREPONTON
1963	AUG				08.0		111.99	3.9	21	0 1	JD	JUAB CO, 12 MI SW OF NEPHI
1963	AUG				02.9	41.54	112,19	3.6	21	DL	JD	BOX ELDER CO, 7 MI NW OF BRIGHAM CTY
1963	AUG				09.8	41.55	112.20	3.5	21	ι	JD	BOX ELDER CO, 12 MI NW OF BRIGHAM CTY
1963	AUG				10.0		111.04	3.9	21	0 1	JD	WASATCH CO, 12 MI NE STRAWBERRY RES.
1963	SEP	02	17	40	06.2	39.03	110.15	3.8		0 (J	CARBON CO, GREEN RIVER
1963	SEP		09	38	45.2	39.62	110.81	2.2		Dι		CARBON CO, PRICE
1963	SEP		20	17	52.9		112.28		2	D L	J	JUAB CO, 2 MI N OF LEAMINGTON
1963	SEP	30	09	17	41.9		111.29	4.5	21	DL	JD	GARFIELD CO 14 MI NE OF BOULDER
1963	NOV	13	06	17	32.8		112.72	3.8				BEAVER CO. 5 MI SW OF BEAVER
1963	NOV	15	05	33	56.4	40.73	112.26	2.8	2	0 (,	TODELE CO, 8 MI W OF MAGNA

YEAR	914	DAY	HR	мN	SEC	LAT-N	1.0º16-W	MAG	INT	DEP	s	D	REMARKS
1963	NOV	21.	11	54	57.0	30.57	111.02	2.0		0	U.		CARBON CO. 13 MI W OF PRICE
1963	DEC				23.9	39.55	111.09	2.1			Ű.		CARBON CO. 16 MI W OF PRICE
1963	DEC	20			23.0	39.3	114.3	3.3		33			UT-NEV BORDER, LEHMAN CAVES
1963	DEC	21			23.0	30,3	114.3	3.3	IV		С		UT-NEV BORDER, LEHMAN CAVES APEA
1963	DEC	22			13.4	39.3	114.3	3.3		33			UT-NEV BORDER, LEHMAN CAVES AREA
1963	DEC	24			09.2		110.36	4.1		0	U	2	CARBON CO, SUNNYSIDE
1963	DEC	25			14.5	39.2	114.2	3.6	IV		С		UT-NEV BORDER, LEHMAN CAVES AREA
1963	DEC	26			04.3	39.75	110.49	3.7		0	U		CARBON CO. 14 MI NW OF DRAGERTON
1965	DEC	28			19.5	39.2	114.3	3.4	IV		С		UT-NEV BORDER, LEHMAN CAVES AREA
1963		28			55.0	30.0	114.7			33	С		UT-NEV BORDER, LEHMAN CAVES AREA
1963	DEC	28	15	50	14.2	30.1	114.1	3.3		33			UT-NEV BORDER, LEHMAN CAVES AREA
1963	DEC	29	64	02	03.8	30.1	114.2	3.5	IV		C		UT-NEV BORDER, LEHMAN CAVES AREA
1963	DEC	29	C4	05	12.2	30.1	114.2	3.4	IV		C		UT-NEV BORDER, LEHMAN CAVES APEA
1963		29			04.1		114.17	4.0		20		С	UT-NEV BORDER, LEHMAN CAVES APEA
1963	DEC				59.2	39.1		3.7	IV		С		UT-NEV BORDER, LEHMAN CAVES APEA
1963	DEC				13.7	39.1	114.2	3.4		33			UT-NEV BORDER, LEHMAN CAVES APEA
1964	JAN				07,9		112.72						IRON CO. SE PANGUITCH LAKE
1964	JAN				21.3		109. *5	3.9		50	U.	2	SW WYOMING, SWEETWATER CO
1964	JAN				34.6		114.13	3.6		30	U	2	UT-NEV BORDER, LEHMAN CAVES APEA
1964	JAN				47.R		114.23	3.5					UT-NEV BORDER, LEHMAN CAVES AREA
1964	JAN				05.1		112.69			20	9	0	BEAVER CO. 8 MI 5 BEAVER
1964	JAN		00		37.3		112.72			20	U	2	BEAVER CO. 6 MI S BEAVER
1964	JAN				17.1		114.13			20		v	UT-NEV BORDER, LEHMAN CAVES AREA CACHE CO, SMITHFIELD
1964	JAN		10		01.0		111.51			20	8	~	UT-NEV BORDER, LEHMAN CAVES AREA
1964	JAN		23		42.1		114.22	3.3		0		9	SALT LAKE CO, MURRAY AREA
1964	JAN		20		52.3		111.91	2.0				0	UT-NEV BORDER, LEHMAN CAVES AREA
1964	FEB		00	28	17.8	37 6	114.26	2,9		15		0	SOUTH OF CEDAR CITY UTAH
1964	FEB				30,1		112.28			20	ň	n	BOX ELDER CO, 15 MI NW OF TREMONTON
1964	FEB		11	13	38.9		112.33			20	ň.	n	BOX ELDER CO, 6 MI NW OF TREMONTON
1964	FEB		13	20	11.8	41 82	112.35			20	ŭ	'n	BOX ELDER CO, 11 MI NW OF TREMONTON
1964	FEB				31.2		111.70	2.2		20	ŭ		MORGAN CO. 4 MI NW OF MORGAN
1964	FEB		20	10	49.7	30 42	114.23	3.7				n	UT-NEV BORDER, LEHMAN CAVES AREA
1964	MAR				24.4		111.91	3.9					JUAB CO, 4 MI SW OF NEPHI
1964	MAR				54.1		112.28			20			SEVIER CO, 18 MI SW OF RICHFIELD
		2											
1964	MAR				52.5		114.21	3.4				D	UT-NEV BORDER, LEHMAN CAVES AREA
1964	MAR				39.1		112,47	2.4		20			MILLARD CO, 5 MI E OF COVE FORT TOOELE CO, TOOELE
1964	APR				54.4	40,52	112.31				U		TOOELE CO, TOOELE
1954	APR	06			45.3		112.42			20			TOOELE CO, 4 MI S OF ST. JOHN
1964	APR	09			39.6		113.98			20			BEAVER CO. SE OF GARRISON
1954	APR	10	21			42.01	112.63	2.3		10 20			UT-IDAHO BORDER, HANSEL MTS JUAB CO, 12 MI SW OF NEPHI
1964	APR	17	20		51.3		111.99			20			MILLARD CO, 20 MI S OF DELTA
1964	MAY	21	10				112.71	1.9		20			JUAB CO, 18 MI SW OF NEPHI
1964	MAY		05	21	05.1		112.15	2.4		20			WASATCH CO 5 MT SE OF HEBER
1964	MAY	11	00			30.23	110.04	2.4		20	ň		WASATCH CO, 5 MI SE OF HERER CARBON CO, 3 MI NE OF CASTLE DALE
1964	MAY				59.0	41.85	111.08			20			CACHE CO, NEWTON AREA
1954	MAY		20		57.0		111.86	2.2		20			CACHE CO, 2 MI W OF RICHMOND
1964	MAY				43.4		112.15	2.3		20	ũ	0	BOX ELDER CO, 12 MI NE OF TREMONTON
1964	MAY		22	45	54.6	42.25	112.14			20	ũ	~	SE IDAHO
1964	JUN		06	44	35.2	39.5	110.3	4.5		33			CARBON CO, PROBABLE ROCKBURST
1964	JUN	06	12	47	1.5	39,94	110.51					D	CARBON CO, S DUCHESNE
1964	JUN		0.0	37	20.6		111.14			20		1	SE IDAHO, E OF BEAR LAKE
1964	JUN		19	44	51,5		111.80			10	Ū		SALT LAKE CO, E SALT LAKE CITY
1964	JUN		19	53	24,3	41.43	113.77	2.5		20	U		BOX ELDER CO, 8 MI NE LUCIN
1964	JUN				21.4	40.87	113.15	2.2		20	U		TOOELE CO, 12 MI NE OF KNOLLS
1964	JUN				30.4	41.27	113,45	3.0				D	BOX ELDER CO, N END NEWFOUNDLAND MTS
1964	JUL		01	11	24.3	41.73	111,50			20			CACHE CO, 17-MI E OF LOGAN
1964	JUL		19	27	35.7	38,35	112.27			20	U		PIUTE CO, 7 MI SW OF MARYSVALE
1964	JUL				13.4	38.89	111.05	3.1		20			EMERY CO, NE OF EMERY
1964	AUG		03	0.9	22.7	41.16	111.46			20			WEBER CO, SE OF OGDEN
1964	AUG	04			23.6	41.53	111.51	3.1		20			CACHE CO, AVON
1964	AUG		15	17	58.6	38,97	110.89	3.2		20	U		EMERY CO. 18 MI NE OF EMERY
1954	AUG		19	55	09.8	41.11	112.92			20			BOX ELDER CO, LAKESIDE MTS
1964		12			52.5	39.42	112.04	3.8		50	0	D	JUAB CO, N SEVIER BRIDGE RESERVOIR
1964	AUG	21	05	34	40.3	37.65	111.89			20	11		GARFIELD CO, 12 MI E OF TROPIC
1964	AUG	24	01	41	17.6		112.28			20			MILLARD CO, 6 MI SE OF FILLMORE
1954	AUG	24			02.9		112.23	3.2		20	U	C	MILLARD CO, 14 MI NE OF FILLMORE
1964	AUG	24			37.2		112.08	3.0		20	U	D	MILLARD CO, SCIPIO LAKE
1954	AUG	28	06	50	46.6	37.0	113.1			33	С		ESE OF ST GEORGE

YEAR	MO	DAY	HR	MN	SEC	LAT-N	LONG-W	MAG	INT	DEP	s	D	REMARKS
								3.3		20			SAN PETE CO, 3 MI E OF MANTI
1964	SEP				37.7		111.59	0.0		10		2	SALT LAKE CO. 10 MT NE SALT LAKE CITY
1964	SEP				25.9		112.30	3.4		20			MILLARD CO, 7 MI S OF FILLMORF
1964	SEP				02.4		112.59			20			BEAVER CO. 14 MI N OF BEAVER
1964	OCT				51.3		111.15	2:8				D	CARBON CO. 14 MI NW OF CASTLE DALE
1954	OCT				23.0		111.41			20			SUMMIT CO, ECHO JUNCTION
1964	OCT				20.7		111.81	4.4				D	CACHE CO, 3 MI S OF RICHMOND
1964	OCT				25.B		111.93			20	13		S IDAHO, FRANKLIN CO
1954	OCT				05.7		112,14			30	U.		SALT LAKE CO, 4 MI N OF RINGHAM
1964	OCT	21	20	04	03.8	41.81	112.68			20	U		BOX ELDER CO, HANSEL VALLEY
1964	NOV	04	08	42	54.3	30,55	110.29	3.2				D	CARBON CO. 3 MI E OF SUNNYSIDE
1954	NOV	04	23	30	37.2		111.00			20			CARBON CO, 6 MI W OF HELPER
1964	NOV		03	08	55.5		111.75			10			CACHE CO, 2 MI E OF RICHMOND
1964	NOV	29	09	31	41.3	30.33	112.10			50			MILLARD CO. 25 MI E OF DELTA
1964	NOV				9.00	40.53	112.20			0			TOOELE CO, 8 MI E OF TOOELE
1964	DEC				13.2		110.94			20	U		GARFIELD CO, 26 MI E OF BOULDER UT-NEV BORDER, LEHMAN CAVES APEA
1964	DEC				03.5		114.24			20		D	CARBON CO, 2 MI N OF SUNNYSIDE
1964	DEC				17.4			2.3		20			EMERY CO, 21 MI E OF EMERY
1964	DEC				27.4		110.8A 112.55	3.1		20			BOX ELDER CO, E SIDE HANSEL VALLEY
1964	DEC				05.2		110.32	3.9				n	CARBON CO, 4 MI E OF SUNNYSIDE
1965	JAN				11.9		110.34	4.5		ñ	U.	0	CARBON CO , 2 MI E SUNNYSIDE
1965	JAN				15.2		113.92	4.1		20	U	D	IRON CO, NE CEDAR CITY
1955	JAN				39.7		111.32	2.5		20		÷.	RICH CO, 20 MI SW RANDOLPH
.965	JAN				20.0		112.15	2.7		20	Ū		BOX ELDER CO, GARLAND
1965	LIAL				55.4	37.49	113.14			20	U		IRON CO 13 MI 5 CEDAR CITY
1965	FEB				43.9		112.10			50			SEVIER CO, 16 MI S RICHFIELD
1965	FEB				33.1	39.84	110.45	2.8		n	U		CARBON CO
1965	MAR	09			34.2		111.30	3.5				D	UTAH CO, 20 MI S STRAWBERRY RESERVOIR
:965	MAR				05.3		110.3	3.9		0	С	_	CARBON CO, SUNNYSIDE
1965	MAR				14.6		112.45	3.4		50	0	D	SEVIER CO. 21 MI WSW RICHFIELD
1965	MAR				41.6		110.31	4.0		0	U	0	CARBON CO, 3 MI E SUNNYSIDE CARBON CO, 3 MI SE SUNNYSIDE
1965	MAR				26.8		110.33	4.3		20		0	UTAH-IDAHO BORDER
1965	APR				43.8		111.62			20			TOOELE CO, GREAT SALT LAKE
1965	APR	02	- 22	35	47.5	41, 09	112.37	1.4					
1965	APR	11	01	39	34.2		112.50			50			TOOELE CO, STANSBURY ISLAND
1965	APR				44.4		110.33	2.0		0	U		CARBON CO. 6 MI SE SUNNYSIDE
1965	APR				33.6		112.85	4.1				0	BOX ELDER CO, N END GREAT SALT LAKE SALT LAKE CO, 11 MI E SALT LAKE CITY
1965	MAY				05.7		111.67	2.7		20 20			SALT LAKE CO, S SALT LAKE CITY
1965	MAY				03.7		111,90			20			IRON CO, 5 MI SW CEDAR CITY
1965	MAY				53.4		113.14			20			UTAH CO, 11 MI S THISTLE
1965	MAT				25.3		111.54	4.1		20	n.	D	MORGAN CO. 12 MI NE MORGAN
1965 1965	MAY				12.6		111.67			20		~	UTAH CO. 7 MI N PROVO
1965	MAY				13.0		111.69			20	U		SE IDAHO, FRANKLIN CO
1965	MAY				04.6		112.67	3.4		20	Ū.		BEAVER CO, 7 MI 5 BEAVER
1965	MAY				22.2	41.17	111.26			20	U		MORGAN CO: 23 MI NE MORGAN
1965	MAY				52.8		112.13	3.5		20			BOX ELDER CO, 10 MI NNE TREMONTON
1965	MAY				37.2		112.12			50			BOX ELDER CO. 6 MI SW BRIGHAM CITY
1965	MAY	28	12	45	22.2.		111.89			20			DAVIS CO. 10 MI SE OGDEN
1965	MAY				21.7		113.31			50	U		BOX ELDER CO, S END NEWFOUNDLAND MTS
1965	MAY				27.7		110.15	3.2				0	EMERY CO, 13 MI SE SUNNYSIDE SW WYOMING, S UINTA CO
1965	JUN				28.5		110.48			20			SALT LAKE CO. E SIDE SALT LAKE VALLEY
1965	JUN				13.0		111.87	2.0				0	SEVIER CO, 6 MI NW RICHFIELD
1965	JUN				39.5		112.12	c.U		20		2	MILLARD CO, 23 MI NE FILLMORE
1965	JUN				37.9					10	U.		SALT LAKE CO. 10 MI NE SALT LAKE CITY
1965	JUN				11.4		111.23			20	ü	n	CARBON CO, 5 MI W SCOFIELD RESERVOIR
. 1965	JUN				44.8		113.85			20	11		MILLARD CO, 16 MI NE PRUESS LAKE
1965	JUN				09.5		110.44	4.1		0	U	0	CARBON CO, 9 MI NW SUNNYSIDE
1965	JUN				39.5		110,40	4.4		0	U	5	CARBON CO, 10 MI N SUNNYSIDE
. 905	304	~ 2				20.000	THOUGH S						

Appendix B

The data in this appendix includes a list of 418 earthquakes which occurred within a 186 mile (300 kilometers) radius of Logan, Utah. This data was made available by

The U.S. National Oceanic and Atmospheric Administration (NOAA)

Appendix B

KEY TO EARTHQUAKE DATA

Source: Source of data is listed as follows:
AEC -- U.S. Atomic Energy Commission BCI -- Bureau Central International de Seismologie, Strasbourg, France
CGS -- Coast and Geodetic Survey
EQH -- Earthquake History of the United States
ERL -- Environmental Research Laboratories
G-R -- Gutenborg-Richter
GS -- U.S. Geological Survey, Denver, Colorado
NOS -- National Ocean Survey
PAS -- Pasadena, California
SLC -- Salt Lake City, Utah
USE -- United States Earthquakes

YEAR, MO, DA: Date

HR, MN, SEC: Origin time

LAT, LONG: Geographic latitude and longitude

DEPTH: Focal depth (km) and depth control factor

- A -- Assigned
- G -- Depth restrained by geophysicist
- N -- Held at 33 km (normal depth), when data not sensitive to depth for a shallow focus
- MAGNITUDES: Body- and Surface- (SURF) wave values as determined by Preliminary Determination of Epicenters program. Authority for other magnitudes and local magnitudes according to source codes.
 - INT MAP: Isoseismal map published and source codes

INT MAX: Maximum intensity

PHENOM DTSUNO: Associated phenomena:

C -- Coal bump or rockburst in coal mine

- D -- Faulting and uplift/subsidence
- E -- Explosion-accidental, controlled or suspected
- R -- Rockburst

- RN: Flinn-Engdahl geographic region
- CE: Cultural effects
 - D -- Earthquake was damaging F -- Earthquake was felt
- Q/S: Quality/number of stations
- MAR DG: Marsden Square
 - DIST: Distance in kilometers between the earthquake location and Logan, Utah.
- NOTE: For additional explanation of this key, contact:

U.S. Department of Commerce National Oceanic and Atmospheric Adminstration Environmental Data Service National Geophysical and Solar-Terrestrial Data Center Boulder, Colorado

SOURCE YEAR	MO DA HR MN	SEC LAT	LONG	DEPTH				NT INT PHEN	OM PN CF G/S	MAR DG	DIST (FM)
							LUGHL				
EQH 1853		66.02 39.766M						v	478 F	120 41	230.
EQH 1880		00.07 42.000N						VT	457 5	156 22	47.
EQH 1880		00.02 40.000N						v	478 5	156 02	100.
EQH 1884		00.02 41.500N						VI	478 D	156 11	60.
EQH 1894		00.02 39.900N						v	478 F	100 93	246 .
EQH 1894		00.02 41.200M						VII	478 0	156 12	65.
EOH 1899		CO.02 41.000N						V	478 5	156 12	87.
EQH 1900 EQH 1905		00.02 39.80CN						VII	47P <u>D</u>	120 92	221.
EQH 1905		00.02 42.400N						VII	133 D 476 F	156 12	250
EQH 1906								V	457 5	156 21	90
EQH 1976		00.02 42.500N						VII	457 S	156 01	100
EQH 1910		00.02 40.800N						VII	460 5	155 19	212
EQH 1913		60.02 41.500M						v	457 5	156 22	30
EGH 1913		00.02 42.000N						VII	457 0	156 22	3".
EQH 1914		00.02 42.000A						VT	478 0	156 01	163.
EQH 1915		C0.CZ 41.800N						V	478 F	156 12	31.
EQH 1915		00.02 41.800N						v	478 F	156 12	150.
EQH 1915		00.02 40.500N						v	037 5	156 04	260
EQH 1916		00.0Z 40.000N						v	478 5	156 01	197.
EQH 1916		CO.02 40.0004						V	(33 F	156 34	:79
EGH 1917		C.07 43.000N						v	457 5	156 31	144
EGH 1920		00.0Z 41.500N						VI	478 F	156 12	27
EQH 1920		00.07 41.500						VI	478 F	156 12	33
EQH 1920		00.02 41.500N						VI	479 F	156 12	33
EQH 1923		00.02 43.600M						v	460 F	156 30	227
EQH 1924		C0.02 42.500N						v	457 F	154 21	86
USE 1929		00.07 43.700					-		467 5	156 71	224
USE 1928		00.02 39.610							678 F	120 91	229
USE 1929		JC.07 42.200							457 F	*56 21	71
USE 1930		01.12 42.600Y						VI	457 5	156 21	115
USE 1931		UL.07 41.500*							AFP F	155 19	212
USE 1930		5 00.02 41.900M							037 F	156 15	707
USE 1935		0 00.07 41.800*							137 F	156 15	285
G-R 1934			112.500		6.	GOPAS	1	SE VIII D	478 C	154 12	63
G-R 1934			112.5004			CCPAS		VIII	478 D	156 10	51
G-R 1934			112.500W			25PAS			478	156 12	63
G-R 1934	L5 06 08 L9	9 49.0 41.750	113.0004	4	5.	SOPAS		VI	478 D	156 13	07
CGS 1942	64 18 05 45	5 42.0 41.500	112.3000	1					478	156 12	40
CGS 1943	02 22 14 20	C 00.C 41.(LC	111.500%	2					479 F	156 11	ηo
USE 1948	02 24 02 3	9 01.1 43.500	V 111.500%	4				VI	457 F	156 31	204
CGS 1949	06 09 09 33	3 22.0 42.500	¥ 110.000 W	4					450	156 20	171
USE 195	01 02 19 53	3 04.0 41.500	112.0004	2				IV	478 F	155 12	33
CGS 1950	0 01 18 01 5		N 110.5604		5.	25PAS		v	478 0	1 = 4 - 1	170
CGS 195	1 (2 25 13 3		N 112.0005						478	156 12	197
USE 195	02 21 17 0		110.000					111	460 5	156 3"	207
USE 195	5 05 24 02 5		9 111.5004					VI	478 F	156 01	143
CGS 1954	02 01 03 3.		N 114. COP						033	15F 34	22F .
CGS 195			N 110.75 m						460	156 2"	141.
USE 195	6 10 03 20 2	1 40.0 41.500	N 110.1004	¥ .				IV	46r F	156 11	146.

SOURCE	YEAR	MO	DA	HR	MM	SEC	LAT	LONG	DEPTH		MAGN	ITUDES		INT	INT	PHENOM	RN CE	210	MAR DG	DIST
	a di Tanadan								(KM)	BODY	SURF		LOCAL	MAP		CTSVNO				(KM)
CGS	1957	[7	21	17	30	12.0	41.500N	113.0000	06(E	478		156 13	102.
CGS	1957					47.0		111.000₩									478		156 51	208.
CGS	1957					41.0		111.0000									478		116 11	209.
CGS	1957	11	03	16	58	0.00	42.500N	111.500W									457		156 21	25.
USE	1957					22.0		111.0004							IV		457 F		156 21	105.
CGS	1958					04.0		112.5000									478		156 12	102.
USE	1958	12				00.0		111.500W							VI		479 F		156 11	143.
BCI	1958	r.2	17	11	57	31.64	39.5"GN	113.0002									478		120 53	
USE	1958	12	01	20	50	48.0	40.500%	112.500W							V		478 D		156 02	152 .
CGS	1958	12	01	22	30	15.0	40.500N	112.500%									478		156 02	152.
PCI	1958	12	02	03	23	15.0	40.500N	112.5009							V		470 E		156 52	152.
CGS	1959	n9	62	00	05	05.0	44.000N	110.000W									AFO F		1 6 4:	. oc.
CGS	1960	64	24	20	36	57.0	43.000N	111.500V									457		116 31	14" .
CGS	1960	05	06	20	28	42.0	39.500N	111.COOW									478		*20 9*	262.
CGS	1960	07	09	21	36	41.0	41.500N	112. COUW									479		156 12	
CGS	1960	07	23	01	31	17.0	42.500N	111.500%									457		154 21	PC .
CGS	194:	07	23	34	10	13.6	42.5ULN	111.500%									457		156 21	ec.
CGS	1960	07	23	07	26	49.0	42.500N	111.5004									457		.54 21	sr .
CGS	1950	07	25	01	30	00.0	42.500N	111.500W				a gran an an an an an a star a star					4 = 7		166 21	FK.
CGS	1960	07				18.0		111.500W									457		156 21	86.
CGS	1960	0 A	C7	16	27	16.2	42.400	111.50CW	049						VI		457 D		156 21	75.
CGS	1960	83	07	19	20	15.1	42.500N	111.420W	C49								4 5 7 F		156 21	pa.
CGS	1960	8.9	10	37	41	35.3	42.5110	111.50CW	018					100000000000000000000000000000000000000	V	/	457 n		:56 21	Fr .
· CGS	1960	0 B	20	07	14	41.7	42.50CN	111.700W	030								4=7		156 21	82.
CGS	1960	6.8	20	97	24	44.2	42.60CN	111.700%	132					101-010-010-010			457		156 21	97.
CGS	1960	0.8	20	3.0	:1	54.3	42.360 N	111.36CW	C41						1	1	457 F		156 21	7.
CGS	1960	(8)	20	10	11	41.9	42.500N	111.600₩	051								457 F		156 21	۶۹.
CGS	1960	08	20	10	13	14.9	42.600N	111.4COW	014								\$57		154 21	çq.
CGS	1960	08	20	11	i1	59.0	42.40CM	111.5COW	049								457 F		156 21	75.
CCS	1960	83	20	14	4é	51.3	42.500	111.6001	029								457 F		156 21	۶4.
CGS	1960	08	21	05	1.4	67.3	42.4.1CN	111.6600	025								457		104 21	77.
CGS	1960	C B	26	01	22	39.6	42.400N	111.200%	032								457		156 01	
CGS	1960	09	12	03	40	57.5	39.100N	111.700%	730								478		120 51	207.
CGS	1961	02	15	09	10	49.4	43.300N	111.400₩	025N								457		156 31	174.
CGS	1961	24	11	08	57	52.7	43.4PCA	111.2008	C33N								457		156 31	
CGS	1961	04	16	05	12	39.3	39.3101	111.500%	03F						V	I	478 0	_	120 91	
CGS	1561	05	05	16	12	20.7	39.6001	110.200%	025N								478 D		120 00	
CGS	1961	05	25	18	28	07.2	42.200	111.º00W	033N								457		156 21	
CGS	1961	10	15		25			111.4004									479		120 53	
CGS	1961	10	16	19	13	\$6.5		111.5068				-					478		120 91	
CGS	1961		17			41.8		1 111.500%									478		100 93	
CGS	1961					46.7		112.500%									478		156 02	
CGS	1962	0 R				43.E		/ 111.0000									479	007	100 01	
CGS	1962			13		28.7		111.900%				5.70PAS		USF			478 D	025	156 11	
PAS	1962			16				112.1066					5.10PAS		۷	1	4-0 P		154 01	
CGS	1962	09		14		13.0		111.8004									478 =	007	176 11	and the second second second
CGS	1962					02.9		111.5004							-		478	900	156 11	
CGS	1962					17.4		110.800							I	V	461 F	010	156 3	202.
CGS	1962					37.0		110.000									478		120 90	
CGS	1962	12	11	10	28	17.5	39.406	V 110.3LUM	033								4//	CUP	120 30	(73.

SOURCE	YFAR	MO CM	HR MN	SEC	LAT	LONG				ITUDES		INT	INT	FHETOM	RN CE	G/S	MAR DG	DIST
							(KM)	RUDA	SURF	OTHEP	LOCAL	MAP	MAX	htsvin				(KM)
CGS	1963	02 25	18 45	16.5	42.600N	119.2004	033	4.30					V		460 F	014	155 29	235.
CGS	1963	03 05	01 30	38.7	42.60CN	111.30CW	033						III		457 F	012	156 21	162.
CGS	1963	03 12	23 47	18.2	39.60CN	110.5009	000							R	478	0(5	120 90	264.
CGS	1953	03 27	07 22	19.4	44.30CM	110.700%	033								459	318	156 40	296 .
CGS	1963	04 64	15 36	26.3	42.JULN	111.2004	0.33								457	21.6	:56 21	70.
CGS	1963	04 15	22 18	22.7	39.500N	110.2004	000							R	478	008	120 20	287.
CSS	1963	F4 16			39.7001	110.500W	000							R	478	C.E	122 90	256 .
CGS	1963		13 33		39.5LUN	110.306W	010							R	479	010	126 90	283.
CGS	1963		05 09			110.000₩									078	516	100 90	
CGS	1963		08 10			112.0004									457	. nr	15F 42	and the second se
CGS	1963		15 51			110.000W									459	618	156 40	
CGS	1963		19 20			111.900%		4.4"					VI		479 [054	120 91	
CGS	1953		9 15 20			111.9008									478	0.06	15, 61	
CGS	1963		20 25			111.2008				-					478	714	156 11	227.
CGS	1963		18 32			111.4004		4.20							478	000	120 01	
CGS	1963		09 45			114.5004							IV		133 E	Crr	156 34	
CGS	1963		12 30			112.200W		3.70					ΙV		478 F	C ^ 7	156 17	
CGS	1963		03 21			112.1000		3.40					IV		475 F	00F	120 92	
CGS	1963		5 07 01			112.200W		3.60							478	0.00	156 12	
CGS	1963		1 05 09			112.2004		3.51							478	162	154 12	
CGS	1963		10 23			110.700W		3.51							470	006	156 01	
CGS	1963		03 15			112.0000		3.50				And the second s			478	006	156 02	
CGS	1963		00 13			111.9008		3.40							479	:	156 1	
CGS	1963		2 17 40			110.1000									478	010	126 9	
CGS	1963		9 18 55 1 23 40			113.300W									457		156 33	
CGS	1963	09 1				112.100₩								E	478	909	156 00	and the second sec
CGS	1963		2 04 37 2 08 58	15.1		111.40CW									457	008	156 31	
CGS	1963		09 56			111.600W									457	07	156 31	
CGS	1963		2 17 06			111.000W									457	068	156 31	
CGS	1963	09 23				111.5058									457	006	156 3	Contraction in the Contraction of the
CGS	1963		2 21 32			111.4004									457	nrp	156 7	
CGS	1963	29 2				111.3000									457	are	156 7	
CGS	1963		3 23 27			111.5004									457		156 7	
CGS	1963		4 17 05			111.200									457	175	156 3	a man in the second second
CGS	1963		1 19 18			111.300%									457	DOR	156 7	
CGS	1963		9 05 58			111.400W									457	1.1.0	156 7	
CGS	1963		9 06 05			111.400%									457	01.9	156 7	
CGS	1963	1. 1		53.1		111.1000									457	017	156 3	
CGS	1963		2 66 58			110.9009									461	91.6	156 3	
CGS	1963		2 21 59			111.1059									457	018	156 3	
CGS	1963	16 1				111.3008		3.90							457	006	156 3	
CUS	1963	10 1		47.1		111.21.6									457	207	156 3	
CGS	1963	10 1	4 08 31	23.0	42.200N	108.3004	030								460	005	155 21	
CGS	1963	11 2		14.5		111.2004									457	:15	156 3	1 157.
CGS	1963	10 2		9 33.0		111.600%		4. * (457	01.5	3	
CGS	1963	15 2	9 17 42	2 11.8	43.200	111.300%	030								057	075	156 3	1 165 .
CGS	1963	1: 3	1 08 18	52.0	43.L	111.3006	033	3.00							457	005	156 3	
CGS	1963	11 0	3 18 26	6 2.0	43.100	111.306%	033								457	DCE	156 3	1 : 4.
CGS	1963	11 0	5 03 44	4 39.7	43.1001	111.200%	033								457	005	156 7	1 157.

SOURCE	YEAR	MO	DA	HR	MN	SEC		LAT	LONG	DEPTH		MAGN	ITUDES		11	NТ	INT	PHENOM	PN C	F 0/5	MAR	DG	DIST
									ineren in in the second	(KM)	RODY	SURF		LOC				DTSVNO				0.0	(KM)
CGS	1763	12	09	61	45	18.2	43	.6001	110.1000	033									460	017	156	3.0	248.
	1963					07.9			110.300%										450	011	156		239.
CGS	1963	12							116.200									R	478	ote	120		287.
CGS	1963	12	26	14	41	06.2			110.1964										478		120		281.
CGS	1964					21.0			109.5004									and the state of the state of	460	000	155		212.
CGS	1964					07.9			111.400%		4.21								457	810	176		167.
CGS	1964					10.4			111.460%										457	007	• • • •		174.
CGS	1964					11.0			111.4024										457	6.00	15,6		174 .
CGS	1964					44.3			111.100%										457	016	11.6		:70.
CGS	1964					28.3			112.3004										457 1		156		47.
CGS	1964					34.8			112.4004										457	010	156		£".
CGS	1964					68.1			112.4000										457	208		22	6'.
CGS	1964					43.1			110.2005										461	000	156		243.
CGS	1964					23.4			111.900		3.90								478	6.5		c1	241.
CGS	1964					49.6			111.400								_	with the second	457	005	156		167.
CGS	1964					31.4			110.8000		3.7:								465	PPE	1 = 6		101.
CGS	1964					02.7			111.5001										457	COF	1 5 4		:51.
CGS	1964					24.1			110.4001														
CGS	1964					35.2			116.807										460	508	156	er	
CGS	1964																						277.
CGS	1964					30.4			110.400										451	UL P			207.
000									111.300										457	005	156		17f .
CGS	1964					49.3			112.100										478	U C E	- in data	12	27.
. CGS	1964					04.3			111.300		3.70								457	006	156		165.
CGS	1964					31.8			1110.3001									R	478	018		10	223.
CGS	1364					59.9			110.2001										47P	007		1 67	297.
CGS	1964					41.9			113.400										478	00.F		17	157.
CGS	1964					15.0			v 111.F00										457	135		21	97.
CGS	1964					25.3			N 111.7001										450	005		2 5 0	156.
CGS	1964					50.9			112.000		3.90								479	ODE		1 00	264 .
CGS	1964					05.1			110.700										459	006		40	296 .
CGS	1964					38.3			112.2001										47P	005		0 92	205.
CGS	1964					35.3			V 111.500										478	0.9		1 51	287.
CSS	1964					20.0			110.000		4.00								46 "	0 n f		r 51	142.
CGS	1964					31.1			113.5001										457	005		5 33	274.
CSS	1964								v 111.300		4.30								470			6 11	15.
CGS	1964					53.8			110.300		4.00							C	479	0.06		0 90	273.
CGS	1964					14.4			v 110.3rc		3.90							p	\$78	0.05		0 0 0	: 77 .
CGS	1965					11.1			V 110.200		4.50							C	478	211		(:77
CGS	1965					34.4			111.300		3.5.1								478	006		0 91	212
CGS	1965					05.3			V 11C.300		3.90							R	478	0.06		0.00	273.
CGS	1965					39.7			× 110.30°		4.00							R	478	017		0 01	297
CGS	1965					24.5			N 110.300		4.31							D	478	0.0.9		0 01	797.
CGS	1965					38.5			N 111.500										457	r 019		F 21	36
CGS	1965	54				51.1			N 111.500										4.57			6 21	01
CGS	1965					25.5			N 111.500		4.50								457			6 21	F.C.
CGS	1965	54				20.0			N 111.500										457			6 21	Pr.
CGS	1965					36.6			112.860										478	015		4 12	of
CGS	1965					25.3			N 111.500		4.10								478	005		6 11	00
CGS	1965					17.6			N 111.400										457	005	1	6 21	121
CGS	1965	05	20	12	1 3	29.2	39	.500	N 110.000	W 533	3.20								478	007	12	û cu	205

$ \begin{array}{cccccccccccccccccccccccccccccccccccc$	URO	CE	YEAR	MO	DA	HR	MN	V SEC	LAT	LONG				ITUDES		INT	TNT	PHENOM	RN	C.E.	015	MAR	DG	PIST
CGS19656627192877.439.4 (*)111.4 (32 / C)4.2 (*)9A 78 <t< th=""><th></th><th></th><th></th><th></th><th></th><th></th><th></th><th></th><th></th><th></th><th>(KM)</th><th>PODY</th><th>SURF</th><th>CTHER</th><th>LOCAL</th><th>**AP</th><th>MAX</th><th>DISAND</th><th></th><th></th><th></th><th></th><th></th><th>(KM)</th></t<>											(KM)	PODY	SURF	CTHER	LOCAL	**AP	MAX	DISAND						(KM)
CGS19656627192877.439.4 (*)111.4 (32 / C)4.2 (*)9A 78 <t< td=""><td>76</td><td>5</td><td>1965</td><td>06</td><td>17</td><td>- 15</td><td>20</td><td>19.8</td><td>39.7116</td><td>111.3000</td><td>1.12</td><td></td><td></td><td></td><td></td><td></td><td></td><td></td><td>679</td><td></td><td>CCF</td><td>126</td><td>01</td><td>234.</td></t<>	76	5	1965	06	17	- 15	20	19.8	39.7116	111.3000	1.12								679		CCF	126	01	234.
$ \begin{array}{cccccccccccccccccccccccccccccccccccc$		-										0.11						P			005	120		269.
$ \begin{array}{cccccccccccccccccccccccccccccccccccc$																					010	120		277 .
$ \begin{array}{c} \cos 1965 & 07 14 03 55 51.4 39.45 60.0 109.9 60.0 733 3.10 \\ - 68 1965 & 07 2.9 42 55.7 43.5 05.0 111.9 60.0 733 4.40 \\ - 66 195 & 07 2.9 42 55.7 43.5 05.0 11.9 60.0 733 4.40 \\ - 67 & 05 1965 & 07 0.2 17 54 3.3 4.6.5 07.1 11.9 60.0 733 3.40 \\ - 66 & 05 1965 & 11 0.1 12 (8.2 2.4 3.45 05.0 111.1 61.0 7.2 0.5 3.40 \\ - 66 & 1965 & 11 0.1 12 (8.2 2.4 3.45 05.0 111.1 61.0 7.2 0.5 3.40 \\ - 66 & 1965 & 11 0.1 12 (8.2 2.4 3.45 05.0 111.1 61.0 7.2 0.5 3.40 \\ - 66 & 1965 & 11 0.1 2 (8.2 2.4 3.45 05.0 111.1 61.0 7.2 0.5 3.40 \\ - 66 & 1965 & 11 0.2 (8.2 2.4 3.45 05.0 111.1 61.0 7.2 0.5 3.40 \\ - 66 & 1965 & 11 0.2 (8.2 2.4 3.45 05.0 111.1 61.0 7.2 0.5 3.40 \\ - 66 & 1965 & 11 0.2 (8.2 2.4 3.45 05.0 111.4 61.0 7.2 0.5 3.40 \\ - 66 & 1965 & 12 0.2 (5.2 4.4 0.0 00.1 13.46 00.0 3.3 \\ - 66 & 0.5 (1965 & 12 0.2 4.1 0.2 (5.2 4.4.6 -0.0 7.1 11.4 61.0 7.2 3.5 4.6 \\ - 66 & 0.5 (1966 & 0.2 11.2 0.3 0.2 4.9 -0.2 1.0 11.4 76 04 0.3 3 3.2 0 \\ - 66 & 0.5 (11 1.4 7.0 5.4 0.4 1.7 76.0 7.3 3.2 0 \\ - 66 & 0.5 (11 1.4 7.5 0.4 0.3 3.5 0 \\ - 66 & 0.5 (11 1.4 7.5 0.4 0.3 3.5 0 \\ - 66 & 0.5 (11 1.4 7.5 0.4 0.3 3.5 0 \\ - 66 & 0.5 (11 1.4 7.5 0.4 0.3 3.5 0 \\ - 66 & 0.5 (11 1.4 7.5 0.4 0.3 3.5 0 \\ - 66 & 0.5 (11 1.4 7.5 0.4 0.3 3.5 0 \\ - 66 & 0.5 (11 1.4 7.5 0.4 0.3 3.5 0 \\ - 66 & 0.5 (11 1.4 7.5 0.4 0.3 3.5 0 \\ - 66 & 0.5 (11 0.4 5.5 0 & 0.6 0.7 111.4 0.0 7.5 3.5 0 \\ - 66 & 0.5 (11 0.4 5.5 0 & 0.6 0.7 111.4 0.0 7.7 0 \\ - 66 & 1.5 & 0.4 1.7 0 & 0.4 1.7 7 \\ - 66 & 1.5 & 0.4 1.7 0 & 0.4 1.7 7 \\ - 66 & 1.5 & 0.4 1.7 0 & 0.4 1.7 7 \\ - 66 & 1.5 & 0.4 1.6 & 0.4 1.4 0.0 7 111.4 0.0 7 0 & 3.7 0 \\ - 66 & 1.5 & 0.4 1.7 0 & 0.4 1.7 \\ - 66 & 1.5 & 0.4 1.7 0 & 0.4 1.7 0 \\ - 66 & 1.5 & 0.4 1.1 0 & 0.4 0 & 0.4 1 \\ - 66 & 0.5 & 0.4 1.1 0 & 0.5 & 0.4 0.7 \\ - 66 & 1.5 & 0.5 & 0.4 1.1 0 & 0.0 111.1 0 & 0.0 7 3 \\ - 66 & 0.5 & 0.5 & 0.5 & 0.5 & 0.5 & 0.5 & 0.5 & 0.5 & 0.5 \\ - 66 & 0.5 & 0.5 & 0.5 & 0.5 & 0.5 & 0.5 & 0.5 & 0.5 & 0.5 & 0.5 \\ - 66 & 0.5 & 0.5 & 0.5 & 0.5 & 0.5 & 0.5 & 0.5 & 0.5 & 0.5 & 0.5 \\ - 66 & 0.5 & 0.5 & 0.5 & 0.5 & 0.5 & 0.5 & 0.5 & 0.5 & 0.5 & 0.$												46.51									000	120		276.
$\begin{array}{cccccccccccccccccccccccccccccccccccc$												3.10									00.	119		300.
$ \begin{array}{cccccccccccccccccccccccccccccccccccc$																					216	156		150.
$ \begin{array}{c} \mbox{CGS} & 1965 & 0.8 & 22 & 17 & 54 & 33.53 & 4^2.5^{0} \mbox{CV} & 11.3 & 1.6 & 0.0 & 173 & 3.547 \\ \mbox{CGS} & 1965 & 11 & 0.8 & 10 & 10 & 0.2 & 10.2 & 0.0 & 113.5 & 0.0 & 0.33 & 3.6 & 0 & 4.57 & 0.55 \\ \mbox{CGS} & 1965 & 11 & 0.8 & 16 & 47.5 & 42.6 & 0.0 & 113.5 & 0.0 & 0.33 & 3.6 & 0 & 4.57 & 0.55 & 0.55 & 0.5 & 24 & 52.0 & 44.0 & 0.0 & 0.33 & 3.6 & 0 & 4.57 & 0.55 & 0.55 & 0.5 & 24 & 52.0 & 44.0 & 0.0 & 0.33 & 3.6 & 0 & 4.57 & 0.55 & 0.55 & 0.55 & 0.5 & 24 & 52.0 & 44.0 & 0.0 & 0.33 & 3.6 & 0 & 4.57 & 0.55 & 0.55 & 0.5 & 24 & 52.0 & 44.0 & 0.0 & 0.13 & 4.00 & 0.033 & 3.6 & 0 & 4.57 & 0.55 & 0.55 & 0.5 & 24 & 52.0 & 0.5 & 0.55 &$												4.000									1.6	156		73.
$ \begin{array}{cccccccccccccccccccccccccccccccccccc$												3 80									005	155		117.
CGS1965110.4121212233050474																					011	156		92.
Cos196511286465Cos19651205062452044Cos196512022410052404000Cos19651202040203040504Cos196512020504041140033.5004Cos19660211200505041140033.5004Cos19660317114757.5040900133.500505Cos196604171955.70409001313.000334.40070																					(12	120		250
$\begin{array}{cccccccccccccccccccccccccccccccccccc$												0.27									206	TEF		112.
$ \begin{array}{cccccccccccccccccccccccccccccccccccc$																					ALE	156		296.
C6319660211203526.0 $42.100N$ 111.400V0333.20 457 e C6319660212095238.542.300N111.200V0333.20 457 e <td></td> <td>7.01</td> <td></td> <td></td> <td></td> <td></td> <td></td> <td></td> <td></td> <td></td> <td>017</td> <td>156</td> <td></td> <td>140.</td>												7.01									017	156		140.
C6S19.660.212295228.6642300N111.260V0.333.2048.748.7C6S1966C317114757.041.776N111.500V(4544.4°478478C6S1966C423202054.539.200N111.400V0334.4°478478C6S1966C423202054.539.200N111.400V0333.44478478C6S1966C4222020.4639.60N111.400V0333.44478478C6S1966C42810.2611.400V0333.408.778.776.77C6S1966C611.9454.7.943.100N111.200V0333.704.57C6S1966C611.9457.1.43.400111.200V0334.57C6S1966C611.9457.1.43.20N111.200V0334.57C6S1966C611.9721.4.243.20N111.200V0334.57C6S1966C611.9721.4.243.20N111.400V0333.404.57C6S1966C611.9721.4.243.20N111.400V0334.574.57C6S1966C611.9721.4.243.20N111.400V0334.574.57C6S1966C611.97<																					190	156		F1.
CGS196603114755.041.776N11.500V(454.4°V $A7g$ CCGS196604170957.540.900V113.300V0334.4°478478478CGS196614 23202054.539.20N111.400V0334.4°478478478CGS196614 24122559.60A111.400V0334.4°478478478CGS196610194539.20N111.400V0333.7U877657CGS19661019457.94.10N111.300V0733.7U857CGS19660611097214.243.200N111.300V033457CGS19660611097214.243.200N111.400V033457CGS19660611097214.243.200N111.200V033457CGS19660611097214.243.20N111.400V033457CGS19660611097243.20N111.400V033457CGS19660611097243.20N111.400V033457CGS196606111057.443.20N111.400V033457CGS1966061110<																					000	154		77.
$\begin{array}{cccccccccccccccccccccccccccccccccccc$																	V				629	156		28.
CGS 1966 C4 23 20 54.5 39.200 11.4000 033 4.44° 978 678 CGS 1966 C4 24 22 59.80 39.2000 11.4000 033 4.44° 678 678 CGS 1966 C4 24 12 26.90 39.2000 11.4000 033 3.70 8.77 8.77 778 6.75 CGS 1966 C4 30 18.26 14.1 15.90 77.00 3.70 8.77 778 6.75 765 11.05 75.91 76.90 3.70 8.57 6.57 6.57 6.57 6.57 75.55 11.05 75.72 8.57 6.57 6.57 75.55 75.55 1.09 72.4.2 4.3.2000 11.1.2000 0.33 3.70 8.57 6.57 6.57 75.55 19.66 6.11 10.97 2.4.2 4.3.2000 11.1.2000 0.33 4.57 6.55 19.66 6.11 10.15 77.4 4.3.2000 11.1.2000 0.33 4.57 6.55 1.56 <td></td> <td>•</td> <td></td> <td></td> <td></td> <td>CCF</td> <td>166</td> <td></td> <td>156.</td>																	•				CCF	166		156.
C63 1926. C4 24 22 59 28.6 33 2.61 4.8 C63 1966 C4 28 12 26 35 2.61 8 4.8 C63 1966 C4 28 12 26 45.5 37.0 8 478 C65 1966 C6 10 19 45 47.9 43.10000 111.10000 33 3.70 85.7 C65 1966 C6 11 97 43.10000 111.30000 33 3.70 85.7 C65 1966 C6 11 97 24.2 43.20000 111.30000 33 45.7 C65 1966 C6 11 97 51.3 34.20000 111.20000 33 45.7 C65 1966 C6 11 0.9 53.43.20000 111.20000 33 45.7 C55 1966 C6 11 0.9 45.31000 111.40000 33 45.7 C55 1966 C6 11 10 <td></td> <td>4-45</td> <td></td> <td></td> <td></td> <td></td> <td></td> <td></td> <td></td> <td></td> <td>010</td> <td>120</td> <td></td> <td>228.</td>												4-45									010	120		228.
CGS 1966 C4 28 10 26 45.5 39.6 CAN 116.3 COV 0.0 C 3.70 R 478 C CGS 1966 C4 30 18 25 14.1 39.6 CAN 117.4 CUX 100 3.3.70 R 478 C CGS 1966 C6 11 35 55 1.3 45.100N 111.100N 033 3.70 457 457 CGS 1966 C6 11 95 55 1.3 45.100N 111.200N 033 457 457 CGS 1966 C6 11 97 24.2 43.100N 111.200N 033 457 CGS 1966 C6 11 07 76 3.43.200N 111.100N 457 457 CGS 1966 C6 11 10 77.4 43.200N 111.200N 33 440 457 457 CGS 1966 C6 11 10 11.400N 11.400N 453 457 457 CGS																					2:7	120		299.
Cos1966(4 3318 2914.1 39.4° Cos 31.4° Cos 3.4° R 476 Cos196606101945 47.9 $43.102N$ 111.4064 033 3.70 457 Cos1966061107214.2 $43.200N$ 111.30674 333 457 Cos1966061107214.2 $43.200V$ 111.30674 333 457 Cos1966061107214.2 $43.200V$ 111.2004 033 457 Cos1966061107214.2 $43.200V$ 111.2004 033 457 Cos196606110727.4 $43.200V$ 111.2004 033 457 Cos196606111052.4 $43.200V$ 111.4004 033 457 Cos196606111052.4 $43.100V$ 111.4004 033 457 Cos1966061110.527.4 $43.100V$ 111.4004 033 457 Cos19660611.1015.20V111.4004 033 457 457 Cos19660611.1015.20V111.4004 033 457 Cos1966070728.220V111.4004 033 457 Cos1966070728.220V111.4004 033 457 Cos19660705.20V111.40																		p			013	120		273.
CGS 1966 06 10 19 45 4.0 111.10000 033 3.70 457 CGS 1956 06 11 05 35 51.3 43.1000 111.10000 033 3.70 457 CGS 1966 06 11 97 24 9.3.2000 111.3000 033 457 CGS 1966 06 11 97 24 9.3.2000 111.2000 033 457 CGS 1966 06 11 97 24 9.3.2000 111.2000 033 457 CGS 1966 06 11 10 15 27.4 43.2000 111.2000 033 457 CGS 1966 06 11 10 15 27.4 43.2000 111.4000 033 457 457 CGS 1966 06 11 10 16 27.4 43.2000 11.4000 033 457 CGS 1966 10 07 3.200 11.40000 033 457 <t< td=""><td></td><td></td><td></td><td></td><td></td><td></td><td></td><td></td><td></td><td></td><td></td><td></td><td></td><td></td><td></td><td></td><td></td><td></td><td></td><td></td><td>010</td><td>120</td><td></td><td>26c.</td></t<>																					010	120		26c.
CGS 1956 65 11 35 35 31 35 35 35 457 CGS 1966 06 11 09 72 14.2 43.200N 111.306W 033 457 CGS 1966 06 11 09 72 14.2 43.200N 111.200W 033 457 CGS 1966 06 11 09 72 43.200N 111.200W 033 457 CGS 1966 06 11 10 74 43.200N 111.100N 33 3.40 457 CGS 1966 06 11 10 57 43.200N 111.100N 33 3.40 457 CGS 1966 06 11 10 57 43.200N 111.400W 33 457 CGS 1966 06 11 15 51 9.40 111.400W 33 457 CGS 1966 07 07 38.200N 111.400W 33 457 457 CGS																					CCF	156		160.
$\begin{array}{cccccccccccccccccccccccccccccccccccc$																		0-10-10-			r	154		167.
CGS 1966 06 11 09 32 49.1 43.100N 111.2004 033 457 CLS 1966 06 11 09 76 3.3 43.200N 111.2004 033 457 CLS 1966 06 11 10 77.4 43.200N 111.2004 033 457 CLS 1966 06 11 10 64 7.8 43.200N 111.4004 033 3.40 457 CLS 1966 06 11 10 15 12.9 43.100N 111.4004 033 457 CLS 1966 06 11 0.9 11.4004 033 457 CLS 1966 06 10 7.3 22.700N 11.4004 033 457 CLS 1966 06 17.0 18.2 Cl 10.400V 033 478 CLS 1966 07 05 18.2 Cl 11.400V 043 478 CLS 1966 07 05 20.02<	0.0																					116		145
C6S19560611097653.343.200N111.202N033457C5S16660611101527.443.200N111.102N033457C5S19660611101527.443.200N111.102N033457C5S19660611101647.843.102N111.402N033457C6S19660614135310.943.102N111.402N033457C6S19660619077822.342.700N111.402N033457C6S1966070512.941.500N110.600N033457457C6S1966070512.041.241.202N101.400N033457C6S1966070520.0241.340.200N109.000N0084.10478C6S1966070520.0241.540.200N109.000N0084.10847C6S1966070525.511.600N0603.40657657C6S19661021.53.9400111.600N0334.104.78C6S196611.421.93.54.600N111.400N0334.104.57C6S1966104.5200N111.400N0333.704.574.57C6S1966																					006	156		157.
CSS19660611101527.443.200N111.102V7333.404.7CSS19660.611104647.843.102N111.402V7333.404.57CSS19660.611104510.0111.402V7333.404.57CSS19760.61110510.943.102N111.402V7334.57CSS19760.61215412.9.141.500N103.402V7334.57CSS19760.610777822.342.702N111.402V7333.704.57CSS19660775200.241.340.200N109.400V0.653.70478CSS19660775200.241.440.200N109.400V0.644.10478CSS1966100.4153.440.200N109.400V0.644.104.78CSS19661010153.940.00N11.400V0.534.574.57CS19661014211935.941.600V11.400V0.534.57CS19661014211935.941.600V11.400V0.534.57CS19661014211935.941.600V11.400V0.534.57CS1966																					nes	156		167.
$\begin{array}{cccccccccccccccccccccccccccccccccccc$												7. 6 1		State and the	Concerns and the second						DOP	156		171.
C6S19760611135310.943.100N111.400U033457C6S197610707822.342.700N111.400U033457C6S197601077822.342.700N111.400U033457C6S1966070510.041.840.200N109.000U006478C6S19660705200241.840.200N109.000U006478C6S19660705200241.840.200N109.000U006478C6S19660705200241.840.200N109.000U006478C6S19661004153435.400110.300U6003.807456C6S19661004153432.20N110.400U6003.807457C6S1966101043200N110.400U6003.807457C6S196610209193.39.400N110.400U63478C6S196610209195.33.960N111.700U63477C6S1966111111455.400N111.700U7533.70457C6S1966111111455.400N111.700U7533.70457 </td <td></td> <td>0.40</td> <td></td> <td></td> <td></td> <td></td> <td></td> <td></td> <td></td> <td></td> <td>OFF</td> <td>156</td> <td></td> <td>152.</td>												0.40									OFF	156		152.
CGS 1966 C6 14 15 01 10.6000 033 4.0 CGS 1966 C6 10 07 38 22.3 42.700 111.6000 033 457 CGS 1966 C6 13.0 46.200 111.6000 053 457 CGS 1966 07 05 12.0 12.41.6000 107.0000 066 3.70 478 CGS 1966 07 05 2.0 12.41.6000 100.2000 006 478 CGS 1966 10 04 15 34 36.2000 107.0000 064 478 CGS 1966 10 04 15 34 36.00 4.10 8 478 CGS 1966 10 04 15 34 36.00 14.600 64 3.80 457 CGS 1966 10 12 95.88 43.6000 11.7004 3.37 457 CGS 1966 11 14 11.1004 03 3.70 <td< td=""><td></td><td>5 C</td><td></td><td></td><td></td><td></td><td></td><td></td><td></td><td></td><td></td><td></td><td></td><td></td><td></td><td></td><td></td><td></td><td></td><td></td><td>CI F</td><td></td><td></td><td>150.</td></td<>		5 C																			CI F			150.
C6S 1976 06 19 67 38 22.3 42.700 111.406 0.53 45.7 C6S 1966 07 05 18 26 10.400 0.050 0.050 47.8 C6S 1966 07 05 12.00 40.200N 109.000 006 47.8 C6S 1966 07 05 2.5 3.8 39.400 102.300 000 4.1 47.8 C6S 1966 07 05 2.5 3.8 49.400 106.300 600 3.80 47.8 C6S 1966 16 16.7 3.8 4.1 6.5 6																					00=	156		106.
CGS 1966 C7 05 18 26 13.0 46.200N 109.000N 005 3.70 478 CGS 1966 07 05 20 02 41.3 40.200N 109.000N 008 478 CGS 1966 10 05 20.02 41.3 40.200N 109.000N 008 478 CGS 1966 10 04 15 34 36.0 10.300N 400 4.10 8 478 CGS 1966 10 04 15 34 36.0 11.600N 600 3.80 10.450N 45.0 CGS 1966 10 18 27 93.80 11.600N 600 3.80 10.500N 45.0 CGS 1966 10 14 21 19 38.3 3.90 10.500N 47.0 CGS 1966 10 12 15 39.500N 11.100N 03 47.0 47.0 CGS 1966 11 11 16 53.60N 11.100N <																					C.C.F.	156		100.
CGS 1966 07 05 20 02 41.3 40.200N 109.000N 006 416 CGS 1966 07 05 20 02 41.8 39.460N 116.302N 000 4.17 R 478 CGS 1966 10 45 55.4 8 39.460N 116.302N 000 4.17 R 478 CGS 1966 10 45 55.4 8 50.001 116.007N 000 3.80 C 450 CGS 1966 11 42 19 55.8 45.200N 111.400N 033 457 457 CGS 1966 11 42 19 55.3 59.600N 111.100N 033 3.70 457 CGS 1966 11 11 45 34.6 39.600N 111.100N 3.3.70 457 CGS 1966 11 11 45 34.6 39.600N 112.700N 3.20 478 CGS 1966 11 12 66 <td< td=""><td></td><td></td><td></td><td></td><td></td><td></td><td></td><td></td><td></td><td></td><td></td><td>3.70</td><td></td><td></td><td></td><td></td><td></td><td></td><td></td><td></td><td>010</td><td>1 6.6</td><td></td><td>204.</td></td<>												3.70									010	1 6.6		204.
CGS 1966 07 30 25 35 46 95 4.1 f R 4.1 f CGS 1966 16 04 15 34 36.0 (41.600) 11.600% 600 5.80 1 457 CGS 1966 10 C4 15 34 36.0 (41.600) 11.600% 600 5.80 1 457 CGS 1966 11 C4 15 34 600% 11.700% 7.00 (7.00) 11.700%												5.70			the later days of Complete and						017	1		294.
CGS 1966 16 04 15 34 36.0 41.600N 110.600N 060 3.80 2.45 CGS 1966 10 CA 15 29 53.8 43.200N 111.60NN 033 457 CGS 1966 11 14 21 93.8 43.200N 111.100N 033 477 CGS 1966 11 21 93.9 600N 111.100N 033 477 CGS 1966 11 21 17 15 11.00N 033 3.70 457 CGS 1966 11 11 16 53.460N 111.10N 033 3.70 477 CGS 1966 11 11 16 53.460N 11.500N 18.20 478 478 CGS 1966 11 14 14 0.52.2 41.700N 112.70%2 473 478 CGS 1966 11 14 14 0.52.2 41.700N 112.70%2 433 478 CGS 1967												6 . 1 C						P			612	120		293.
CSS 1966 10 CB 15 29 53.8 43.200N 111.60NN C33 457 CGS 1966 11 14 21 19 36.9 41.600N 111.70NN (C F 457 CGS 1966 10 14 21 19 36.9 41.600N 111.70NN (C F 457 CGS 1966 10 19 15.3 39.600N 111.100N 033 3.70 457 CGS 1966 11 11 16 45 34.600N 111.700N 15.20 478 CGS 1966 11 11 16 45.900N 112.700N 133.3 478 CGS 1966 11 14 14.900N 12.700N 133.70 478 CGS 1967 01 05 22.42 01.40 112.700N 133.70 478 CGS 1967 01 05 22.42 01.40 11.2.700N 133.70 478 CGS 1967 01 05																the rest of the break					PLE	116		103.
CGS 1966 11 12 19 36.9 41.600x 11.700x 600x 11.100x 63.5 CGS 1966 10 20 99 19 15.3 39.600x 11.100x 63.5 67.7 CGS 1966 12 717 15 11.100x 63.3 67.7 65.7 CGS 1966 11 11 11.6 45.200x 11.110x 63.3 67.7 CGS 1966 11 11 11.6 45.200x 11.500x 61.7 3.20 47.8 CGS 1956 11 12 66.7 47.8 47.8 47.8 47.8 CGS 1956 11 14 14.30 52.2 41.700x 112.700x 63.3 47.8 CGS 1967 01 05 52.42 01.4 41.848x 112.5570x 035 47.8 CGS 1967 01 05 52.42 01.4 41.848x 112.5570x 035 47.8 CGS 1967 01 05 <td></td> <td>- e (1</td> <td></td> <td></td> <td></td> <td></td> <td></td> <td></td> <td></td> <td></td> <td>DCE</td> <td>156</td> <td></td> <td>177.</td>												- e (1									DCE	156		177.
CGS 1966 10 20 99 15.3 39.600N 111.100V 033 47P CGS 1966 10 27 17 15 11.0 43.200W 111.00V 033 457 CGS 1966 11 11 16 53.4 3.400 10.500V 015 3.20 478 CGS 1966 11 12 66 41.900N 112.700V 033 478 CGS 1966 11 14 12 52.24 41.700N 12.700V 033 478 CGS 1967 01 05 22.42 01.4 0.1640V 11.2700V 033 478 CGS 1967 01 05 22.42 01.4 0.6650 478 CGS 1967 01 05 22.42 01.4 0.1640V 11.925V 0.0565 478 CGS 1967 01 0.11.7 39.640V 11.925V 0.33V 478																		٢			225	106		C7.
CGS 1966 10 27 17 15 11.0 43.200% 111.000% 733 3.70 457 CGS 1966 11 11 16 45.34.6 25.6600% 11.500% 15 3.20 478 CGS 1966 11 12 66.74.9 41.900% 13.27.00% 033 478 CGS 1966 11 14 43.52.22 41.77.0% 033 478 CGS 1967 01 05 22.42 01.4 41.648% 112.557% 035 478 CGS 1967 01 05 22.42 01.4 41.648% 112.557% 056 478 CGS 1967 01 05 22.42 01.4 41.648% 12.255% 478 478 CGS 1967 01 05 22.42 01.4 11.929% 833% 478 CGS 1967 01 05.22 42.01 41.733% 476					_		-														nne	190		249.
CGS 1966 11 11 11 11 12 66 10 11 15 67 67 CGS 1966 11 12 26 76 41.90000 112.70000 033 478 CGS 1966 11 14 14 10 52.2 41.7000 112.70000 478 CGS 1966 11 14 14 10 52.2 41.7000 112.70000 478 CGS 1966 11 14 14 10 52.2 41.7000 112.70000 478 CGS 1966 11 14 14 52.4 41.7000 112.70000 478 CGS 1966 11 14 11.11 12.9000 112.70000 478 CGS 1967 11 23.0000 40.0000 478 478 CGS 1967 11 2.10000 41.97000 478 CGS 1967 11 2.21000 4.11.920000 478 CGS 1967 10 2.21000 4.71000 478												3.70									007	156	-	177.
CGS 1956 11 12 76 12.700 M 033 478 CGS 1966 11 14 10 52.2 41.700 M 12.700 M 033 478 CGS 1967 01 05 22.42 01.44 112.700 M 033 478 CGS 1967 01 05 22.42 01.44 114.5457 W 0055 478 CGS 1967 01 05 22.42 01.44 114.5457 W 0055 478 CGS 1967 01 02 14.71 39.640M 111.923W 033 478 CGS 1967 01 22.14 12.723 M 033 478																					DOF	120		26F .
CGS 1967 61 65 24.700 112.700% 63.3 478 CGS 1967 61 05 22.42 61.4 41.668% 112.557% 6056 478 CGS 1967 61 05 22.42 61.4 41.668% 112.557% 6056 478 CGS 1967 61 22.14 61.61 73.9 476																					DDE	156		82.
CGS 1967 81 05 22 42 81.4 41.6488 112.5578 0856 478 CGS 1967 61 22 10 01 01.7 39.6408 111.9298 0338 479																					207	156		73.
CGS 1967 fl 22 10 fl E1.7 39.640M 111.929W H33N 47R	CO	GS																	478	2	833	156		61.
																					010	120		237.
CGS 1967 C2 05 10 J7 16.6* 39.549N 110. 972 C33N 3.61 478	CO	GS	1967									3.61							471		FOF		01.	286.
	C	55	1967														V	1			0.24		ro	206.
	C	GS	19F7																47	p	015	156	13	136 .
																					010	150		244 .
	Ċ	GS	1967																47	R	DC.6	150	11	10.
	C	US	1967									3.70						г	46	0	016	156		101.

SOURCE	YEAR	MO	DA	HR	Mt	SEC	LAT	LONG				ITUDFS		INT	INT PHENOM	RN CE	015	MAR	DG	DIST
									(KM)	BODY	SURF	OTHER	LUCAL	MAP	MAX DTSVVO					(KM)
CGS	1957	63	05	05	41	23.9	41.3341	111.669₩	011	3.50					IV	478 F	002	156	11	50.
CGS	1967					35.4		110.238₩		3.70						460	r11	156		135.
CGS	1967					1 59.8		111.8994							IV	478 F	007	156		110.
CGS	1967	63	03	21	17	12.0	39.493N	111.1184	664	3.41						478	211	.20	0 1	267.
CGS	1967	14	08	21	42	13.8	42.869N	111.3674	233N	4.20						457	010	151	21	130 .
CGS	1967	04	24	17	.07	7 28.4	41.535N	110.6189	0006	3.40					E	AFO		156	11	114.
CGS	1967	05	01	07	37	7 47.3	43.512N	111.7084	033V	3.20						457	0.05	156	31	194 .
CGS	1967	06	26	22	31	02.8	43.C00M	111.0004	033	4.00						457	006	156	31	157 .
CGS	1067	07	21	15	21	7 58.0	41.3CON	113.4COW	019							47 P	61 "	156	13	141 .
CGS	1967	14	24	11	53	3 53.2	42.0001	111.3°0W	033							457	069	156	21	133.
CGS	1967	09	02	10	- 6.4	4 07.6	41.100%	111.6004	006							478	uut	156	11	77.
CGS_	1967	0.9	11	04	_1	0 46.5	43.000N	111.000W	033	3.50						457	009	156	31	157 .
CGS	1967	09	11	14	21	9 1.6.4	43.000N	111.2000	033							457	006	156	31	146 .
USE	1967	09	24	34	41	6 48.8	40.7001	112.140W				3.70			V	478 D		156	12	:21 .
CGS	19F7	09	24	05	01	0 20.1	40.70CN	112.100W	064	3 . 7 1						478	607	15F	12	*21 .
CSS	1967	10	25	00	5	6 12.2	39.50CN	117.490%	0.00	3.10						478	nr s	120	511	270.
CGS	1967	10				3 00.4	39.500N	110.300W	000	3.50						478	013	120	01	· · · · ·
CGS	1967	10	25	02	1	7 45.9		110.402W		3.50						478	010	120		270.
CGS	1957					1 34.4		110.300W		4.95						479	03 F	120		267.
CGS	1967		25			3 (110.300W		4.00						470	610	120		·
CGS	1967		31					111.5000								457	007	11.6		of .
CGS	1967					6 31.9		111.700W								47F	DC.E	120	A of States Party of	286 .
CGS	1967					9 48.1		116.200W		3.90						478	013	15(277.
CGS	1967					6 54.7		110.6004								459	005	156		200.
USE	1967			13				111.706W		4.30						478 5	017			57.
CGS	1967					5 43.2		111.800								478	012			10.
CGS	1967			22				109.8000								460	01E	155		246 .
CGS	1967					3 23.6		110.4009								450	0.10			254.
CGS	1968		14			9 22.4		111.6008		3.90						457	014	156		116.
CGS	1968					8 44.0		112.10LW		4.1^						478	078	_120		275.
CGS	1968					7 52.3		112.1004		3.90						478	014			275.
CGS	1968					0 12.1		112.100W										120		275.
CGS	1965					9 54.1		111.960%								479	015	126		286.
CGS	1963					2 54.2		112.000		4.00						478	006			286 .
CGS	1968	01		04				112.2004		2.80						479	015	120		276.
CGS	1968	- 12	19	-				112.100H								457	6.07	120		115.
CGS	1968			0 06		4 30.2		110.600		3.70						460	677			106.
CGS	1968	02		10				111.0004		3.20				and the second sec		478	010			251.
CGS	1968					4 29.7		110.3004		2.00						478	010	125		201.
CGS	1968		6			7 12.5		112.7001								478	60.8	156		74.
CGS	1968		21			8 12.6		113.300%								478	TTA			144.
CGS	1968			1 01		3 48.8		111.3351								4 . 7	200			76 .
CGS	1968					9 28.0		116.3514		3.80						478	012			280.
COS	1968		0			1 12.5		111.1444					-			478	019			25P.
USE	1968		0			3 36.4		111.400;		4.00					TI	478 F	613			
CGS	1968					7 04.7		110.9901								479	272	120	cr	267.
CGS	1768					1. 28.7		110.948:								478	009	120	00	241.
CGS	1968					1 48.1		110.2485		4.20					C	470	211		5-	207.
CGS	1968	69	1	0 6	0 5	2 53.5	* 42.582N	111. 1991	1 604							457	006	1 . 6	21	102.

SUURCE	YEAR	MO DA HR	MN SEC	LAT	LONG	DEPTH		MAGNI	ITUDES		INT	INT P	HENOM	RN CE	0/S	MAR DG	DIST
						(KM)	PODY	SUPF	OTHER	LOCAL	MAP		TSVNO				(KM)
CGS	1020	16 11 55	56 17 7	70 500	110.3849	0110	3.20							478			
CGS		11 16 03			110.3848		3.90							AFC	009	120 90	
CGS	1968		33 37.5				4.FU							478	022	120 50	
CGS	1968	12 18 11			110.8600		3.50							459		156 41	
CGS	1969		55 38.4*											457		156 21	
CGS	1969		11 19.7		111.1494		3.60							457	007	156 3	
CGS	1969	02 28 15			111.8119									457	01.0		
CGS	1969	03 13 07	03 14.8	39.444N	110.2268	002	4.10							478	011	120 91	
CGS	1269	04 05 10	56 35.4	43.195N	111.321%	C33N								457	0.7	156 3	
CGS	1969	16 30 12	05 52.3	42.690N	111.1690	033N	3.70							457	012	156 2	1 116.
USE	1969	08 27 15	59 28.4	42.900%	110.5000	3016	4.21			4.10CGS		III		161 F	012	156 2	152.
CGS	1969	08 27 18	35 18.9	43.002N	110.7244	0015	3.90			3.90065				460	011	156 31	
CGS	1969	(8 30 02	01 02.5*	43.672N	110.6754	0016	3.90	0.0000000000		3.60065				450	005	15£ 3	177.
CGS	1969		31 45.9		111.4204		4.10			4.40065				457	310	156 3	
CSS	1959		33 15.0		111.4298		4.5?			4.00065				457	714	150 2	1 14(.
CGS	1969		57 18.7				4.30			4.40055				457	C14	156 3	
CES	1969			42.964N			3.90			4.10055				457	2**	156 2	
CGS	1969		12 16.7*				3.1"			3.80055				457	LU L	and a second second second second	the summer of the state
CGS	1969		58 13.5				3.90			3.80095				457	010	156 2	1 130 .
CGS	1969		19 45.0*							3.ences				457	006	15F 2	the house and have been seen and
CGS	1969		13 51.€				3.90							4 = 0	010		
CGS	1970		49 23.0				4.00						r	479	313	120 9	and and the second second second second second
CGS	1970			39.350N			4.10						С	478	r67		
USE	1970		40 41.2				4.F.					<u>v</u>		478 D	018	156 1	
CGS	1970		46 54.2				4.20							47R	61 0	120 9	
NOS	1970		06 33.3				4.40							460	619		
NOS	1970		46 42.1*				4.30			3.00NOS				478	514	156 2	
NOS	1970		11 51.5				4.30			2.60.005				478	009		A RANNEL MALLON AND A
NOS	1970		45 45.1*				3.90			2.00000				418	010		
1.05	1971		UP 22.8				3.34			3.60NOS				478	010		
NOS	1971		1 03.2							3.4 0105				478	0.07	7. 5	
FRI	1971		01 50.5+											450	0.05		
FPL	1971		22 37.2*				3.80							479	CC7		
ERL	1971		54 19.7				3.60							457 F	PFF		
ERL	1971		35 12.3.				4.10							250	110	156 2	
ERL	1971		44 57.6				4.20							450	012	156 2	
ERL	1972	03 06 13	33 24.5*	41.873*	111.6161	005G	4.50							478 F	015	1 77 1	1 21.
ERL	1972	10 01 15	42 25.2	40.581N	111.3021	6955	4.70							479 0	025	156 (1 130.
ERL	1972	16 16 21	49 32.4	40.415*	111.0391	0110	4.10							479 5	009	156 0	1 164.
ERL	1972		5 3E 06.9				4.41					ĪV		457 E	r14	156 2	
ERL	1973		57 53.0											457	008		
ERL	1973		5 26 64.54											4 f 5	n n 7		
FRL	1973		56 21.5.											459	803		and the local of the second second
ERL	1973		5 38 43.0.											450	006		
ERL	1973		40 26.3.											450	CLE	4	201.
ERL	1973		4 36 25.24											45.0	006	156 4	
LRL			9 16 49.9											450	108		
ERL	1973		0 30 58.7				3.60							45.0	007		
ERL	1973	04 13 06	56 51.3	42.0901	11c .t 43	1 2116								457	007	156 2	77.

SOURCE	YEAR	"0 D	A HR	N.N.	SEC	LAT	LONG				ITUDES		THT		FHENOM	PN CF	015	MAR	ns	
								(KM)	BODY	SURF	OTHEP	LOCAL	MAP	MAX	ntsvrc					(KM)
ERL	1973	04 1	4 JE	45	49.2	42.051N	112.525 W	624	4.40			4.75ERL		IV		457 F	036	156	22	66.
ERL	1973	04 2	0 15	06	54.1+	4".853N	111.0F3W	[17								457	504	156	71	240.
ERL	1973	16 1	4 21	59	41.6*	42 . P 31 N	168.5348	0006							F	461	C.L			295.
GS	1973	C7 1	6 56	36	42.8*	39.1491	111.50 SW	(115	4.21							478	LIF	120	01	202.
GS	1973	10 2	7 00	44	07.9*	42.770N	111.109%	611								457	007	156	21	126.
GS	1973	11 2	2 23	36	31.05	42.1421	112.6924	6106								457	011	156	32	72.
GS	1974	07 0	4 (3	10	56.2*	44.4141	111.112W	OUEG								158	G	156	41	300.
<u>65</u>	1974						111.053W							III		458	F	156	41	295.
65	1974						111.1374									458	7	156	41	2 91 .
GS	1975						112.5454		4.40			4.2. SLC				457 F		156	22	۴P.
GS	1975						112.548%		6.11	6.0		6.27PAS		VIII		457 0	258	156	22	68.
65	1975						112.4818		4.30			3.00SLC				457	16	156	22	63.
35	1975						112.534W		4.17			3.Prelc				457	36	156		66.
GS	1975						112.453		4.31			3.27560				457	2.0	156		۴٦.
GS	1975						112.5218		4.75			4.73SLC				457 F	pr	156		64.
6 S	1975						112.578W		4.30			4.10SLC			and the second	457	45	* 56		۴۶.
GS	1975						112.5200		4.00			2.93SLC				457	E	15F		72.
GS	1975						112.605.4		4.31			3.40SLC				457	24	156		71.
GS	1975						112.644		3.90			2 . 82 SLC				457	7	156		77.
C S	1975						112.537%		4.00			2.61SLC				457	E.	154		66.
GS	1975						112.5949		4.00			3.23SLC				157	13	156		6°.
GS	1975						112.605		4.00			3.60SLC				457	32	156		70
55	1975						112.5005		4.30			3.ERSLC				457	27		22	6 .
GS	1975						112.497		4.40			3.00510				457	7		22	62
GS	1975						112.411		4.50			3.19SLC				457	8		1.2	E G .
GS	1975						112.444		4.70			3.2°SLC				457				. 63
GS	1975						112.4774					2.90SLC				4 5 7	P		20	55
GS	1975						112.4931					3.20510				457	17		53	62
GS	1975						112.4926		4.66			3.10SLC				457	21		- 22	E.
GS	1975						112.585					3.10SLC				457	20		22	76
GS	1975						112.4984		4.40			3.00510				457	20		5 22	63
GS	1975						112.374%		4.70			2.95510				478	5			47
GS	1975						112.554					3.20SLC				457	22		5 22	<u>E</u> E
55 65	1975						108.770		3.2							460	14	155		253
							108.864		3.7							460	c		5 18	251
65	1975						114.003		4.3							037	23		6 35	192
GS	1975						112.466		0.0			7 00010				457 F	16		6 22	65
	1975											3.00SLC		11		460	7		Transfer and	summer and the local
CS GS	1975						112.448		3.3			3.30ERD 3.60SLC				457	74		6 30	21P 65
65	1975						112.448					2.50580				457	12		6 41	274
65	1975						112.571					4.00SLC		111		457 F	24		F 22	76
65	1975						112.489					4.00300		111		457	2 4 R		6 22	
55	1975						112.426							III		478 F	23		6 12	51
GS .	1975						112.4531		4.2			3.60\$LC		T		457 5	74		6 22	
GS	1975						114.716		4.1			CRECCEL		1 4		037	6		6 14	
65	1975						114.533									037	11		6 04	
GS	1975						114.096					3.7PERD				C 37	F.		6 -4	
55	1975						112.556									457	E		6 22	6.6
GS							112.517									45.7	- 7	15		

								-	1000	1000	1010 1000	LUCAL					(KK)
S	1975	11	0 []	0 19	16.3.	41.822	11 11 00 19 16.3* 41.822N 116.595W 010	Ju nos	9 1					465		156 1	11.9.
S	1975	11 1	17 C	8 21	11.10	1 41.955	11 17 C8 21 11.1U 41.955N 112.533W C07	334 CG	1			3.P0SLC		457	33	154 12	
5	1975	12 0	12	1 30 1	40.5	44.417	44.417N 111.4434 1116	13 121	31			2.40ERD		0 L 0	12	156 4	1 20
S	1975	12 2	12 20 01		44 12.9		42.031N 112.5284 6856	CRW CO	50		2.70ERC			457	11	156 2	s.
-	1975	12 2	213	1 53	*8°.	43.154	19.8* 43.154N 111 .752U PUER	13 125	54			3.10EPC		460	c	5 935	1
GS	1976	02 1	11 03	3 28 1	14.70	1 41.271	14.7U 41.27"N 111.8434	210 M24	2			2.70SLC	111	47P F	22	1 5 3 1	
1	1976	02 1	4	3 11	11.0.	42.719	11.0* 42.719N 111.272W 0056	724 06	956			3.40AFC		457	-	154 2	11
	1976	02 2	1 1	4 12	12 47.5	41.589	41.589N 112.5534 0056	53 W 90	950			3-3-AEC		457	G	1 7 1 1	
10	1976	02 2	23 0	101	14.2.	42.610	23 01 01 1P.2* 42.01PM 112.5234 0056	23× 00	50			2. RLAEC		457	1	154 2	2
	1976	02 2	27 0	7 18	16.4 +	41.24	41.24UN 111.275W	75 W CC	0050			2.41SLC		4 7.0 F	11	1 231	1
6 S	1976	63 6	0 10	7 23	11.1	42.159	07 07 23 11.1 42.159N 112.5714 (714 CO	0-0			of all the same and the second second second second second		457	12	156 2	2
5S	1976	03 1	15 0	2 28	34.5.	+ 43.253	02 28 34.5* 43.253N 110.732W SC56	32W 30		3.7		3.20AEC		462	. 1	156 3	10
	91:1	13 1	17 0	P 23	46.3.	* 43.243	17 3P 22 46.3* 43.243N 110.6P4W 0050	00 M 4d		3.9		7.5CAEC		460	1.1	156 3	
	1976	53	10	1 52	33.31	+ 43.258	33.3* 43.258N 110.692W PCF6	3 M26	95					455	11	156 3	
	19161	03 2	21 0	8 03	07.3.	43.272	21 0R 03 07.3+ 43.2724 110.6P34 017	P3W 01	1					451	11	156 3	-
	1976	23	22 0	19 16	45.7.	+ 42.051	18 45.7* 42.050N 112.6474 BUEG	00 626	156					457	4 I	156. 2	0
	1976	6.6	14	9 37	57.81	U 42.118	57.8U 42.118N 112.49CW	ACW 907	17			3.4PSLC		45,7 F	30	116 2	
	1976	30	15 0	2 08	15.41	U 41.896	15.4U 41.886N 112.442V	100 X24	I.			3.1°SLC		4-16	ц. с.	1 7 2 1	