# Chapter 4 Geophysical Survey (SIP, IP Methods)

### 4-1 Outline of the Survey

#### (1) Purpose of the Survey

The survey refers to an area where a mineralized zone of porphyry copper type has been found through geological and geochemical surveys of the initial phase. In the second phase, semi-detailed geological, geophysical and drilling surveys were successively carried out aimed at exploration of the mineralized zone. The SIP survey lines were survyed for the purpose of determining continuity of the mineralized zone between three drill holes.

This year, the geophysical methods of SIP and IP were used to unravel emplacement condition and continuity of the mineralized area.

#### (2) Area of the Survey

The area of the SIP survey is situated around the confluence of Maden Stream and Hasan Stream, some 4 km southeast of Hamusiköy Village. Seven survey lines 200 m apart were laid parallel with Line B connecting holes MJT-2 and MJT-3 drilled in the second phase survey. The area and arrangement of the survey lines are illustrated in Figs. 2 and 9.

## (3) Survey Specifications

Field work specifications were set as follows:

a. Electrode Configuration dipole-dipole array

b. Electrode Separation 100 m

c. Electrode Separation coefficient  $n = 1 \sim 5$ 

d. Measurement Method Frequency domain

e. Frequencies SIP 0.125 Hz~ 88 Hz(18 frequencies)

IP 0.125 Hz/ 1.0 Hz

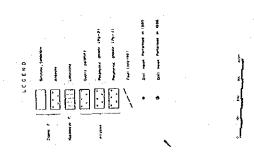
f. Length of Survey Line 18.0 km in nine lines

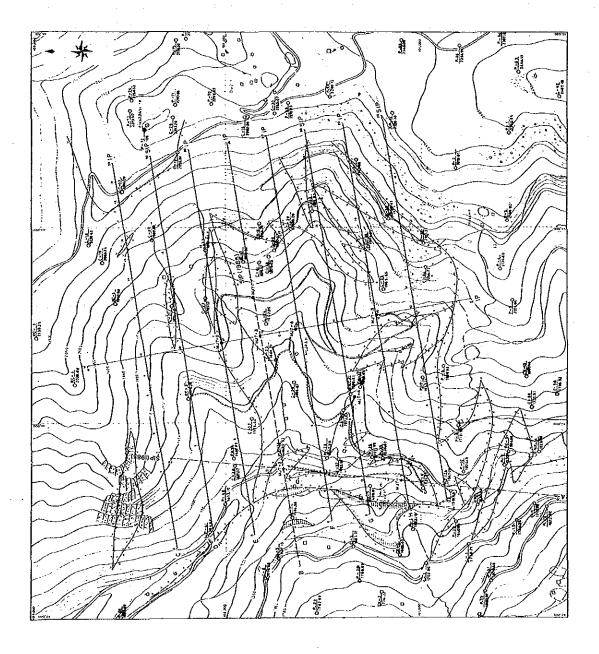
SIP: Line D,G,I 6.0 km in three lines 240 points

IP: Line B,C,E,F,H,J 12.0 km in six lines 480 points

## (4) Survey Method

The SIP method is the abbreviated name of Spectral Induced Polarization and operates on the same principal as the conventional IP method. The SIP method





Location Map of IP and SIP Survey Lines in the Güzelyayla Area 6 Fig.

measures apparent resistivity and phase difference over a frequency range of 0.01 Hz to 100 Hz, while the conventional IP method measures a difference in apparent resistivity expressed in percentage of two frequencies. The measurement data are expressed in spectral diagrams of phase and magnitude, and in Cole-Cole diagrams. Analysis of these responses gives discrimination of minerals or types of mineralization and eliminates electromagnetic coupling which occurs at low resistivity in the ground, at wide electrode separations, and with a large number of electrode coefficients.

In this survey, the Harmonic System of Zonge (USA) was applied. The IP responses over a range of 0.125 Hz to 88 Hz are measured through calculation and extraction of high frequency, using the Fast Fourier Transform of 3rd, 5th, 7th, 9th and 11th harmonics from three fundamental frequencies of 0.125 Hz, 1.0 Hz and 8.0 Hz.

Observation of a wave form is necessary for measurement of phase, and a communication cable which connects transmiter with receiver was laid down parallel with the survey line separated by 25 m to 30 m. At the receiving station, response is amplified through three porous pot-electrodes in copper saturated solution with a copper rod. Amplified responses are transmitted through a communication cable to the receiver (GDP-12/2GB). Data is processed and printed out.

### (5) Measuring Equipment

The equipment used in this survey are listed in Table 1. The illustrated diagrams of the equipment for IP and SIP measurements are shown in Fig. 10 and Fig. 11 respectively.

#### 4-2 Data Processing and Rock Sample Measurement

### (1) IP Data Processing

Panel Diagrams of percent frequency effect (PFE) and apparent resistivity (AR) were provided from pseudo-sections of each line. Three plan maps were prepared on each electrode separation coefficient of n = 1, 3, 5.

Table 1 Measuring Equipment for SIP and IP Surveys

I tem	Spec	ification	Quantity.		
Zonge GGT-5 Transmitter		: 250, 500, 750, 1.000 V	1		
	Output Ampere	: 1.2 - 20 A			
	Have Form	: Square wave			
	Frequency	: 1/8 - 2,048 Hz			
	Weight	: 61 Kg			
Zonge XMT-12	Frequency	: 1/8 - 2,048 Hz	2		
Transmitter controller	Height	: 58 Kg			
	Power Supply	: 12 V Battery	·		
3 7110 K P		e v.			
Zonge ZMR-5 Engine generator	Output Power	: 5 Kw	1		
	Output Voltage	: 115 V			
W 1 0 400 B	Frequency	: 400 Hz			
Honda C-400 Engine		: 10 hp 4 cycles	<del></del>		
Zonge GDP-12/2GB Receiver	Input	: 2-Channel	2		
	Frequency	: 1/8 - 2,048 Hz			
	Senstivity	: 0.2 μγ			
	Weight	: 15 Kg			
	Power Supply	: 12 V Battery			
Zonge CAP-12 Mini Cassette	1	: 6.2 Kg	2		
Recorder	Power Supply	: 12 V Battery	· · · · · · · · · · · · · · · · · · ·		
Tektronic 212 Oscilloscope		· ·	1		
Zonge 180/1 Isolation Amp.	Weight	: 1 Kg	3		
Zonge FP-1 Feild Pressp.	Gain	: 1, 10	3		
D)	0 4 51		900 1		
Electrode	Current Electrode	: Stainless steel	200 rods		
	Potencial Electrode	: Cu-CuSO4 non-polarizable Porous Pots	10 pcs.		
Cable	Current	:	10 Km		
±	Communication	: 640 m length	3 rolls		

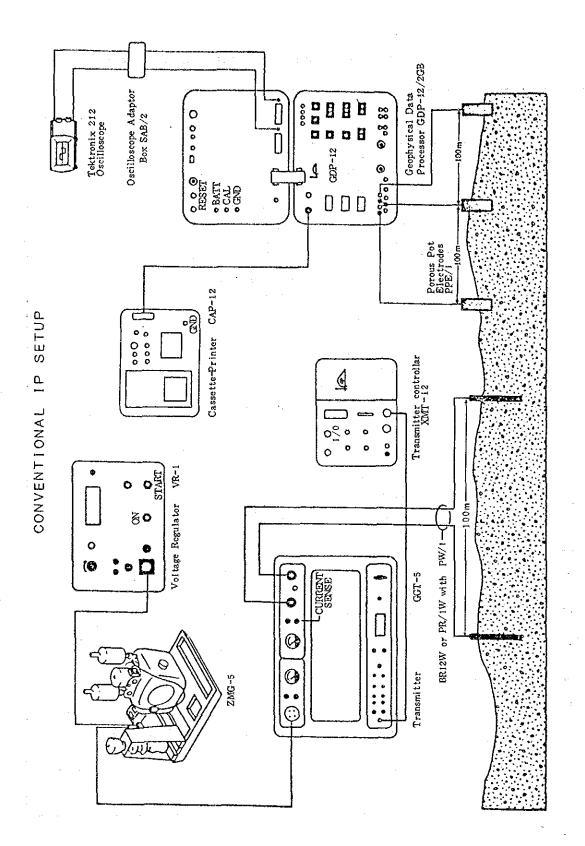


Fig. 10 Illustrated Diagram for IP Equipment

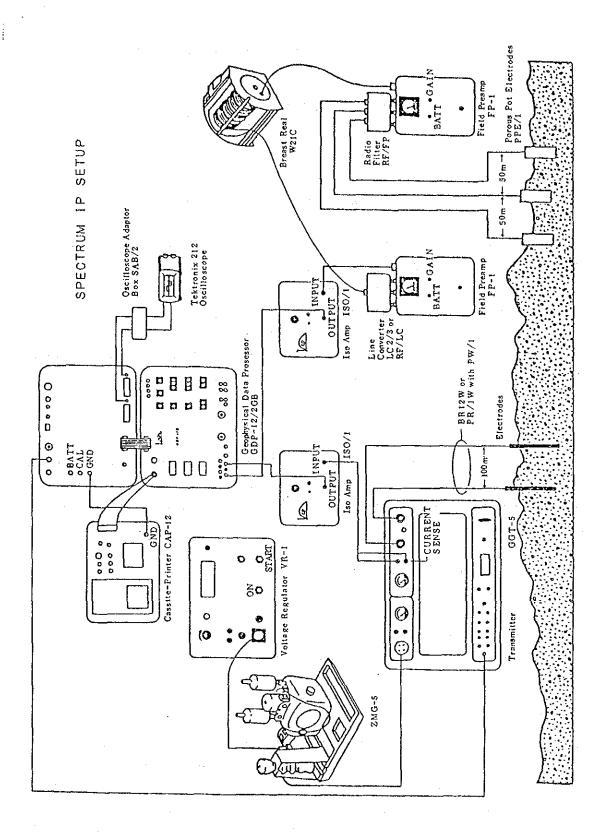


Fig. 11 | Hiustrated Diagram for SIP Equipment

### A) PFE

A value of PFE is calculated by magnitudes (M) at  $0.125~\mathrm{Hz}$  and  $1~\mathrm{Hz}$  as follows:

PFE = 
$$\frac{M(0.125 \text{ Hz}) - M(1.0 \text{ Hz})}{M(0.125 \text{ Hz})} \times 100$$
 (%)

## B) AR

A value of AR is calculated by the following equation:

 $AR = \pi \ a \cdot n(n+1)(n+2) \cdot V/I \qquad (ohm-m)$ 

where, a : electrode separation in meters

n : electrode separation coefficient

V: voltage received in volts

I : transmitted current in amperes

In the present survey, the apparent resistivity at 0.125 Hz was calculated and topographic correction was made with conductive paper.

## (2) SIP Data Processing

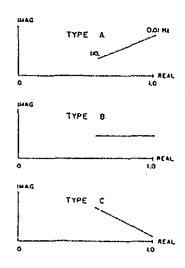
Data obtained in the field consist of real and imaginary parts of complex resistivity response at each frequency, apparent resistivity, phase and magnitude of received basic frequency, and so on. The following figures are determined using these data:

- (1) Cole-Cole Diagram
- (2) Magnitude Spectrum
- ③ Phase Spectrum
- (4) Raw Phase at five frequncies
- ⑤ PFE Pseudo-section
- 6 Apparent Resistivity Pseudo-section

Data processing and method of analysis are given as follows .

## A) Cole-Cole Diagram

In a Cole-Cole diagram, print-out data for each frequency are plotted on a coordinate by setting the negative imaginary part on the vertical axis and the positive real part on the horizontal axis. An example is shown in Fig. 12.  $\theta$  i and Mi on the figure are, respectively, called phase angle and magnitude. The Cole-Cole diagram is known to display a special spectrum depending on the kind of mineral or rock.



According to Zonge et al, there are three types of spectra as illustrated in the left- hand figure.

Type A, showing a pattern of ascent to the right, indicates existence of sulphide minerals, graphite or strong alteration. The flat line pattern of Type B indicates moderate alteration, and the Type C pattern of descent to the right indicates weak alteration, alluvium sediment, fresh igneous rock or limestone. Discrimination of Cole-Cole diagrams of this survey was based on this classification of the three types.

# B) Magnitude Spectrum

The magnitude refers to Mi and Mj of Fig. 12, and is easily obtained from positive real and negative imaginary components of field data. The values are normalized by dividing by Magnitude  $M_0$  of minimum frequency (0.125 Hz). A magnitude spectrum figure is plotted by setting the magnitude value on the vertical axis and frequency on the horizontal axis. (Fig. 13) In the figure, a flat line indicates fresh rock without mineralization or alteration, whereas the spectrum line descending to high frequency indicates strong alteration, sulphide minerals and graphite.

### C) Phase Spectrum

In a phase spectrum, the vertical axis is phase angle  $\theta$  of Fig. 12, and the horizontal axis is frequency. (Fig. 14 a). Data obtained in the field survey are a combination of original IP responses (solid line A in Fig. 14 b) and pseudo-IP responses (dotted line B in Fig. 14 b ) derived from electromagnetic coupling. Line C (-x---x-) in Fig. 14 b shows the combined IP responses. The phase spectrum indicated in Fig. 14 a was obtained through measurement.

### (3) Decoupling Manipulation

Decoupling denotes the removal of a false component in IP responses originating from electromagnetic coupling. The decoupling process was conducted on data over the whole lines of A and B. The decoupling procedure on the SIP measurement in this area was based on the method provided by P.G.Hallof and W.H.Pelton. The analytical method is summarized below. A complex impedance ZA(f) obtained from the SIP survey is approximated by the following equation.

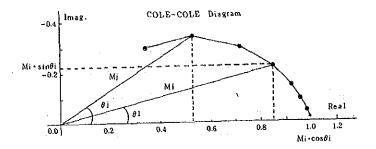


Fig. 12 Cole-Cole Diagram

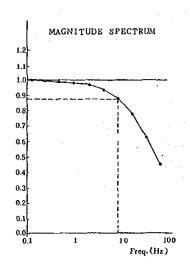


Fig. 13 Magnitude Spectrum

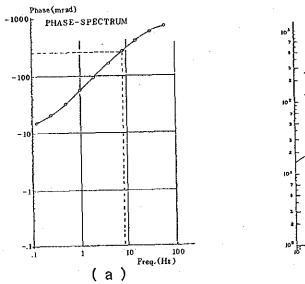


Fig. 14 Phase Spectrum

$$ZA(f) = Ro[1 - m_1 \{1 - \frac{1}{1 + (i2\pi f \tau_1)c_1}\} - m_2 \{1 - \frac{1}{1 + (i2\pi f \tau_2)c_2}\} + m_3 \{1 - \frac{1}{1 + (i2\pi f \tau_3)c_3}\}]$$

where, m; chargeability

 $\tau$  ; time-constant

c; frequency dependence

f; frequency

The equation can be separated into three parts as follows:

$$1 - m_{1} \left(1 - \frac{1}{1 + (i2\pi f \tau_{1})c_{1}}\right) \qquad (1)$$

$$- m_{2} \left(1 - \frac{1}{1 + (i2\pi f \tau_{2})c_{2}}\right) \qquad (2)$$

$$+ m_{3} \left\{1 - \frac{1}{1 + (i2\pi f \tau_{3})c_{3}}\right\} \qquad (3)$$

The first nominal refers to an IP response, the second indicates electromagnetic coupling derived from a homogenous earth and the third represents the value of electromagnetic coupling in a conductor. Ten parameters (Ro,  $m_1$ ,  $\tau_1$ ,  $c_1$ ,  $m_2$ ,  $\tau_2$ ,  $c_2$ ,  $m_3$ ,  $\tau_3$ ,  $c_3$ ) of the equation above are determined from the SIP measurement using the least squares method of a non linear type. The nominals (2) and (3), being the values of electromagnetic coupling, are removed from the equation, and only the complex impedance Zco(f) of the IP response is obtained.

$$Z_{co}(f) = [1-m_1\{1-\frac{1}{1+(i2\pi f \tau_1)c_1}\}]$$

# (4) Rock Sample Measurements

In the analysis and interpretation of the survey results, it is essential to understand the SIP features of main rocks and ores distributed in the surveyed

area. The measurement of SIP was conducted over samples totaling 35 pieces - 25 pieces of drill core in the Hasandere area and 10 pieces of core samples in the karadag area - to investigate spectra of phase and magnitude, Cole-Cole property, percent frequency effects and resistivities. The procedure of measurement is as follows:

① Sample preparation : A cube of 3 cm is prepared.

② Saturation with water: The samples are soaked in distillied water for 24 hours.

③ Measurement

: Instruments used are illustrated in Fig. 15. Except for the laboratory transmitter, all instruments and measuring methods are the same as those used in the field. Standard value of current was set at 50  $\mu$  A.

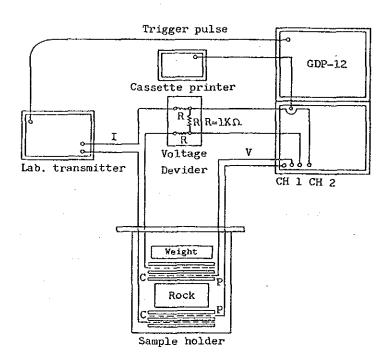


Fig. 15 Laboratory Equipment for Rock Samples

#### Results of Rock Sample Measurements

Results of the laboratory measurements are shown in Table 2 for total samples, and in Table 3, which includes 35 samples from 1985, for each rock type. The results from these samples can be categorized by their phase spectra which fall into seven types as A, B, C, D, E, X and Y which are shown in Fig.16.

Table 2 Results of Core Sample Measurement (Güzelyayla Area)

No.	Location	Rock	Phase	PFE	Resistivity	Phase	Minera-	Alteration
	(m)		(-mrad)	(%)	(ohm-m)	spectrum	lization	
1	16.5	Pgl	31.1	4.3	797	В	Ру	ser-bio
2	52.0	Pgl	83.1	13.3	990	D	Cp, Py	bio-ch
3	100.0	Pgl	8.5	0.4	360	Е		ser-bio-ch
4	150.0	Pgl	41.8	6.3	1,390	D	Ру	bio-ser
5	200.0	Pgl	17.7	2.3	258	D		ser-ch-bio
6	250.0	Pgl	28.7	3.7	852	Y		bio-ser
7	300.0	Pgi	19.5	2.4	1,100	Y,(E)		ser-bio-ch
8	49:0	Pgl	63.0	9.3	3,560	A	Cc,Cp	ser
9	99.0	Pgi	41.2	6.6	2,160	D		ser
11	13.3	Pgl	27.9	3.8	1,440	В		ser
13	100.05	Pgl	18.7	2.5	892	D,(E)		ser-ch
16	250.0	Pgl	27.6	5.2	889	D		
18	16.0	Pgl	16.4	2.7	7,300	D´ ·		ser,Qz vein
19	55.0	Pg1	54.5	10.2	2,530	D		ser
20	255.0	Pg1	47.9	8.3	1,600	D		ser,Qz vein
21	300.0	Pg1	62.7	8.7	4,200	Y	•	ser-anhydrite
22	36.5	Pg1	24.0	3.7	6,700	A		ser
24	146.35	Pg1	20.3	2.7	911	E		ser,Qz vein
25	190.1	Pg1	31.2	3.9	195	E		ser,Qz vein
(Ave	rage value	)	35.0	5.28	2,000			
12	49.8	Pg2	24.7	3.6	1,630	В	Ру	
14	150.0	Pg2	17.5	2.6	3,720	В		
15	198.8	Pg2	40.1	6.0	1,580	В		
17	301.0	Pg2	20.5	2.4	405	E		silicified
(Ave	rage value	)	25.7	3.65	833			`
			•					
10	195.0	BA	114.0	20.2	5,690	D,(B)		
99	46.6	. 02	13.8	2.2	3,120	A		
23	40.0	Qz	10.0	4.4	0,140	Ŋ		

Abbreviation

Pgl: Porphyritic granite

Pg2: Porphyritic granite (Intrusive)

BA: Basaltic andesite

Qz : Qurtz vein

Py : pyrite

ser: Sericite

Cp : Chalcopyrite

bio: Biotite

Cc : Chalcocite

ch : chlorite

Table 3 SIP Response in the Classification of Rock (Güzelyayla Area)

Rock	Sample No.	Phase	PFE	Resistivity	Phase spectrum type						
		(-mrad)	(%)	(ohm-m)	Α	В	C	D	Е	Х	Y
		P									
Andesite											
Andesite	13	-4.3~56.16	-0.19~144.57	126~10,068	4			4		3	2
		(109.6)	(25.68)	(1,753)							
		(10.1)%	(1.73)	(1,976)*	<u>.</u>						
Basaltic	6	3.5~376.2	0.45~88.31	$547 \sim 7,164$	2	1		1		2	
Andesite		(108.5)	(22.20)	(3,970)							
		(14.3)%	(2.19)%	(5,408)*							
Pyroclastic	3	3.9~5.1	0.58~0.73	1,819~4,996	3						
Andesite		( 4.4)	(0.63)	(3,828)							
Porphyritic g	ranite										•••••
Pg-1	28	7.8~208.8	0.40~38.71	195~7300	5	4		11	3	2	3
		(38.5)	(5.97)	(2,281)							
Pg-2	5	17.5~40.1	2.40~ 6.00	405~6,551	1	3			1		
Qurtz Porphyr	y										
	1	6.5	0.84	5,207			1				
Qurtz vein				•							
	1	13.8	2.20	3,120	1						
Calcareous mu	dstone										
: 	1	10.1	1.16	4,322			1				
Siltstone											
3	2	1.9~ 2.4	0.32~0.34	3,273~12,649	2						
Total	60				18	8	2	16	4	7	5

<sup>( )</sup> Average value

<sup>※</sup> Excepted anomalous value

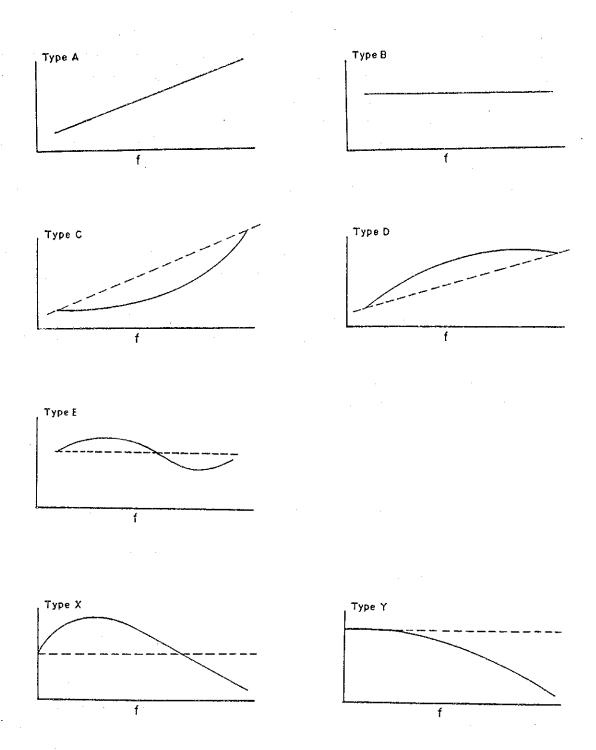


Fig. 16 Phase Spectrum Types of Rock Samples

In comparison, types in the Cole-Cole diagram are divided into two groups; one belonging to type C in the Cole-Cole diagram contains A,B,C and D types of phase spectra and the other belonging to type A in the Cole-Cole diagram contains E, X and Y types of phase spectra.

The result indicates the following:

- (1) Phase and PFE values of andesite and basaltic andesite are the highest. The next is porphyritic granite. In the same rock, these values vary extensively in range depending on condition of alteration and pyrite dissemination.
- ② Phase and PFE are proportionately correlated.
- (3) Resistivity values are in a wide range from 126 ohm-m to 12,650 ohm-m.
  Most values are generally high, but 10 samples (17 % of the total 60 samples) are below 500 ohm-m.
- (4) In the case of the phase spectrum, samples of high phase and PFE belong to the X and Y types. Resistivity is not correlated with the phase spectrum.
- (5) On the other hand, weakly mineralized or fresh rock samples are higher than 2,000 ohm-m in resistivity, and mostly belong to A and D types of the phase spectrum.

#### 4-3 Results of Interpretation

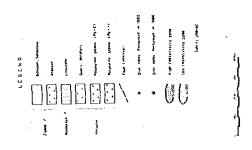
Three plans of AR and PFE were provided for each electrode coefficient of n= 1,3 and 5. Pseudosections of each line were illustrated by a panel diagram. Terrain correction on AR was made with carbon conductive paper to eliminate an effect caused by topographic undulations. As for the SIP responses, phase variation at five frequencies ranging from 0.125 to 3 Hz, phase spectrum, magnitude spect-rum and Cole-Cole diagram were shown on a pseudosection of each line. Phase variation after decoupling was also presented.

### (1) Plan Map and Pseudo-Section of Apparent Resistivity

The AR in the area ranges from 10 to 4,217 ohm-m and gives the arithmetic mean (M) of 269 ohm-m. The standard deviation ( $\sigma$ ) after common logarithms stands at 0.476, giving the values of M+ $\sigma$  and M- $\sigma$  of 804 and 90 ohm-m respectively. Reference values of high and low resistivity were set at nearest contour values of 1,000 and 100 ohm-m. Due to a broadness of AR values, contours were drawn at 10, 20, 50, 100, and so on, to give a uniform interval under logarithms.

# Plans of AR

N=1 Plan (Fig.17): A broad distribution of low AR of less than 100 ohm-m was



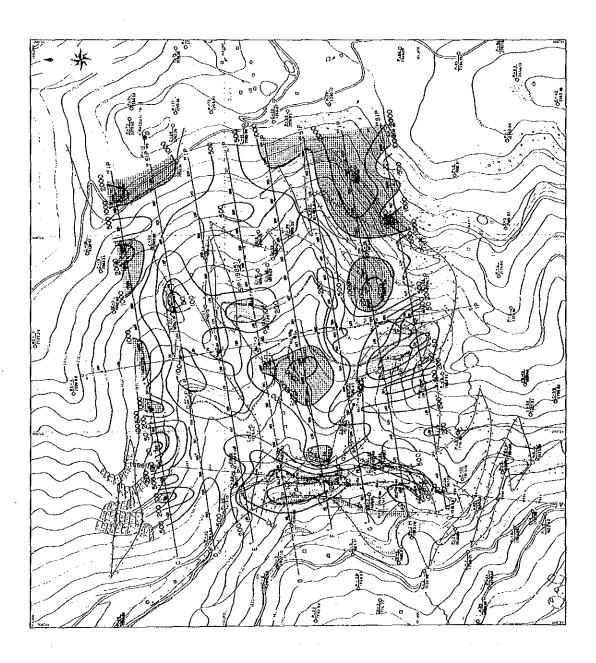


Fig. 17 Plan Map of Apparent Resistivity [0.125 Hz] (n = 1)

delineated in two places in the west at Maden River (in the vicinity of the boundary between andesite and porphyritic granite) and at Hasan River (in an area of andesite). A small zone of low AR is observed in the north.

The most prominent high AR, being more than 1,000 ohm-m, is widely distributed in the southeast (andesite and porphyritic granite Pg-2). Smaller zones are scattered in the central and northeastern parts of the area.

N=3 Plan (Fig.18): The two zones of low AR at Maden and Hasan Rivers on Plan N=1 merged to form a broad area in the southeast (mainly in an area of the porphyritic granite, Pg1). An area of medium scale of low AR is observed in Line E and west of line B.

A zone of high AR is detected at the same place as on N=1, but is smaller in scale.

N=5 Plan (Fig.19): The zone of low AR has a similar pattern with that of N=3, but the zones on Line E and in the middle of line B have a tendency to become wider.

The zone of high AR decreases its size and is detected at three places in the southeast, northeast and northwest.

#### Section of AR (Fig. 20)

Distribution of zones of high and low AR to depht is examined on the panel diagram.

The zone of high AR appears in three places at:

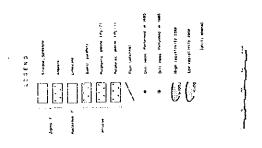
- (a) east (andesite) and west (andesite and quartz porphyry) of Lne C
- (b) in shallow depth between Nos.6 to 9 (porphyritic granite Pg1) on Line F.
- (c) on Line H and east of Line H (andesite and porphyritic granite Pg1 & 2).

The first and the last give continuous patterns to depth and the pattern in (b) is assumed to reflect an effect of high AR at shallow depth.

The zone of low AR is observed in four places at:

- (d) east of Line C
- (e) at the western ends of lines E, B, F, H, (andesite and porphyritic granite Pg1)%
- (f) the centers of lines F and B (rock types as above)
- (g) Line H and the section from No.6 to No.9 (andesite) on Line I which was detected in Hasan river.

The zone of low AR of (e) is situated along Maden river and the zone of (g) is located on Hasan River. Most of them coincide with alteration zones delineated on the surface. No alteration zone is known in relation with (d) and (e) which are situated in topographic depressions. These are probably related



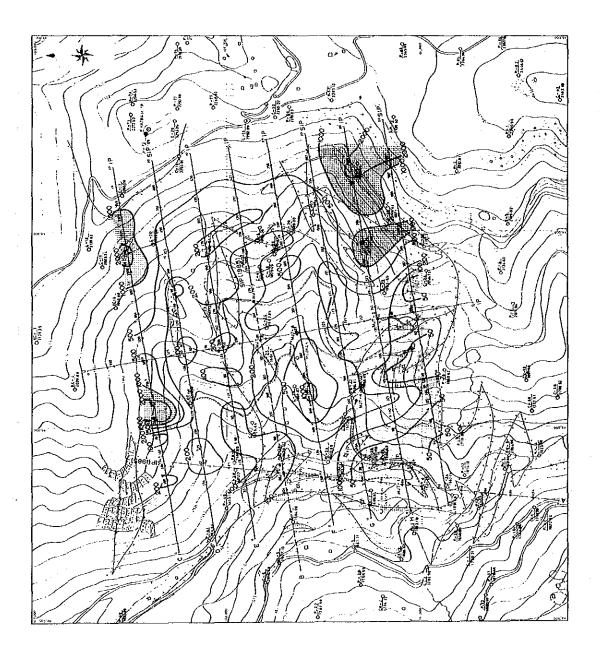
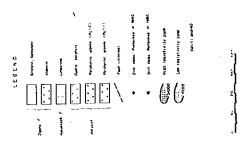


Fig. 18 Plan Map of Apparent Resistivity [0.125 Hz] (n = 3)



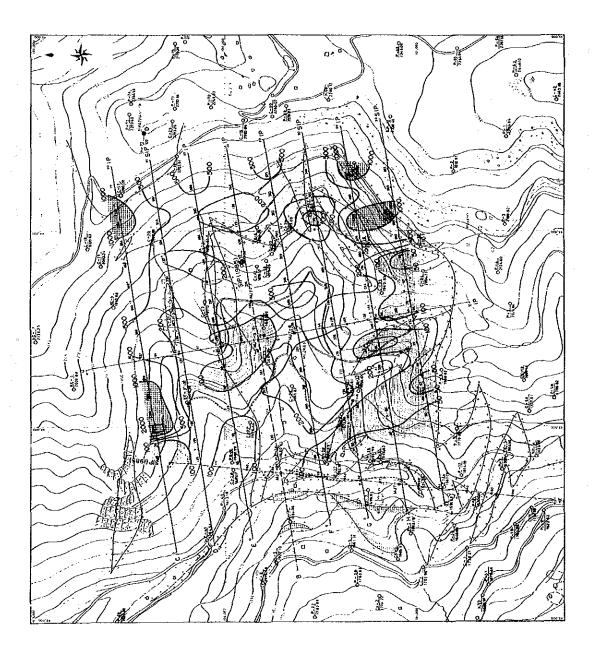
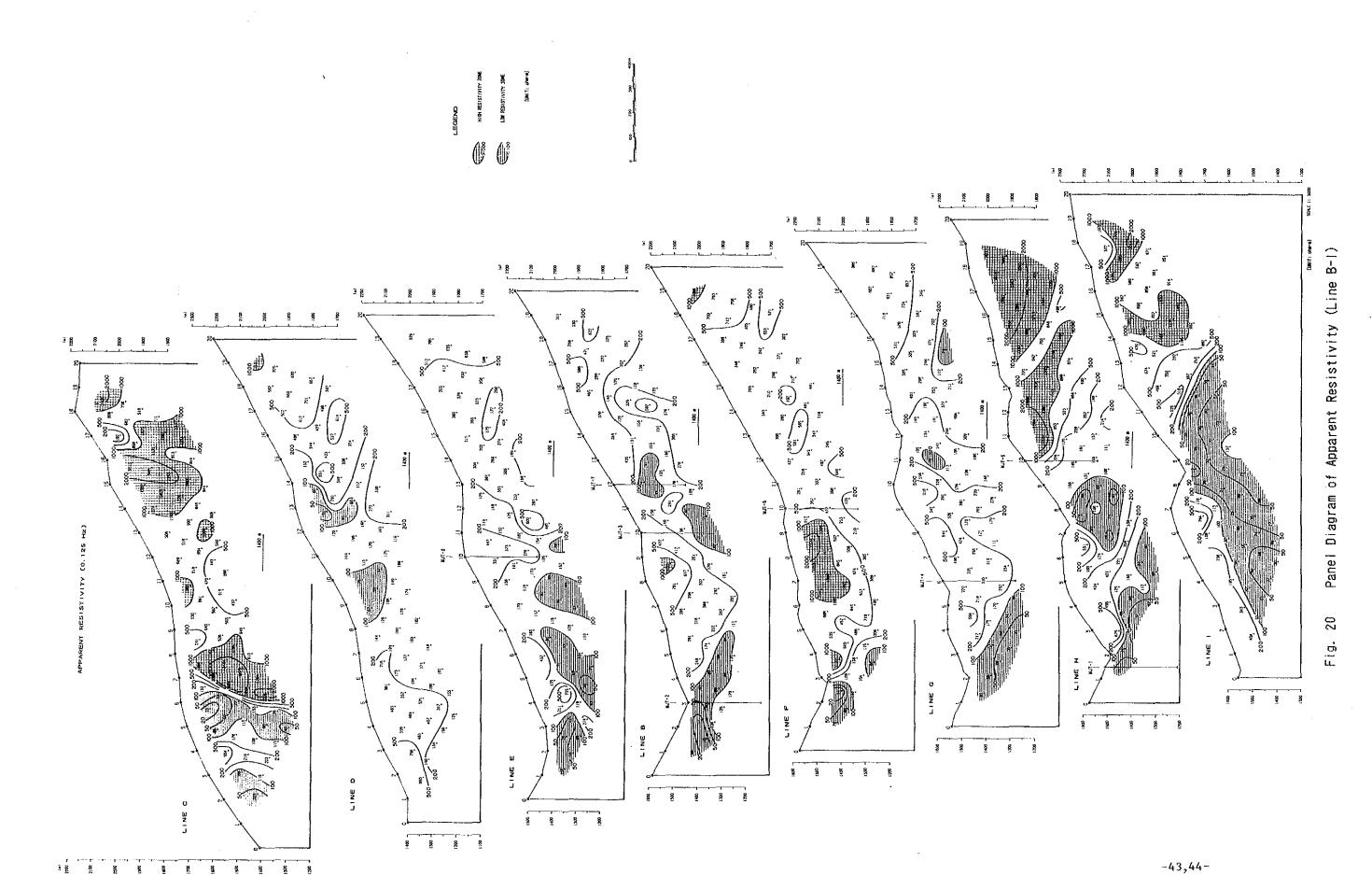


Fig. 19 Plan Map of Apparent Resistivity [0.125 Hz] (n = 5)



with ground water to some extent. The zones of low resistivity are situated at shallow depth except the sections between No.7 and No.9 on line E and between No.2 and No.5 on line I.

The characteristics of the resistivities in the area are low compared with the results of laboratory measurements. They range from 200 to 500 ohm-m of the medium range. The distribution of AR cannot be related with a specific rock but the zone of low AR can be deemed to be related to alteration zones or ground water.

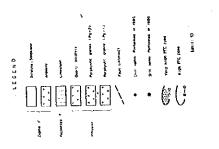
## (2) Plan Map of Pseudo-Section of Percent Frequency Effect

The PFE value of the area ranges from -0.3 to 18.3% and the arithmetic mean (M) stands at 2.720, giving M+ $\sigma$  and M- $\sigma$  at 10.12 and 4.68% respectively. A histogram at the 2% interval gave a logarithmic normal type distribution. Measurements of rock property indicate that PFE values over 4% are anomalous, although 90 % of field measurements exceed this level. Consequently, most of the area can be deemed anomalous. Of these, strongly anomalous areas with a PFE value over 10% are discussed below.

#### Plans of PFE

N=1 Plan(Fig.21): A prominent area of PFE above 10% is observed west of Lines C, D and E and extends from the middle of Line C to west of Line E in a direction of NE-SW. A high anomaly south of Line A, which was detected in the previous year, can be observed at the western end of Lines G, H and I but does not extend towards the north or east. Small anomalies of high PFE are found at No.14 of Line E, at the eastern end of Line B, at No.11 of Line F and at No.17 of Line H. The first two run in a line of NW-SW direction with a zone of more than 8% PFE and stretch to the middle part of Line C. This anomaly forms, with the anomaly of NE-SE direction of Line C, a horseshoe shape opened to the south. In the middle of this, there exists another anomaly which extends from the vicinity of drilling site MJT-3 southward.

N=3 Plan (Fig.22): The zone of high PFE over 8% can be divided into two areas. The first extends from the north, then in a NE-SW direction, and has a broad zone of more than 10% PFE in the north where andesites occur and it becomes smaller on Line B where porphyritic granites (Pg1) exist. Another is a zone located in the east with an elongation of N-S trend. An area of more than 10% at the eastern end of Lines D,B and E over andesites and porphyritic granites (Pg1) is conspicuous. The central part of Lines H and I on the area of porphyritic granites (Pg1) is also distinguished.



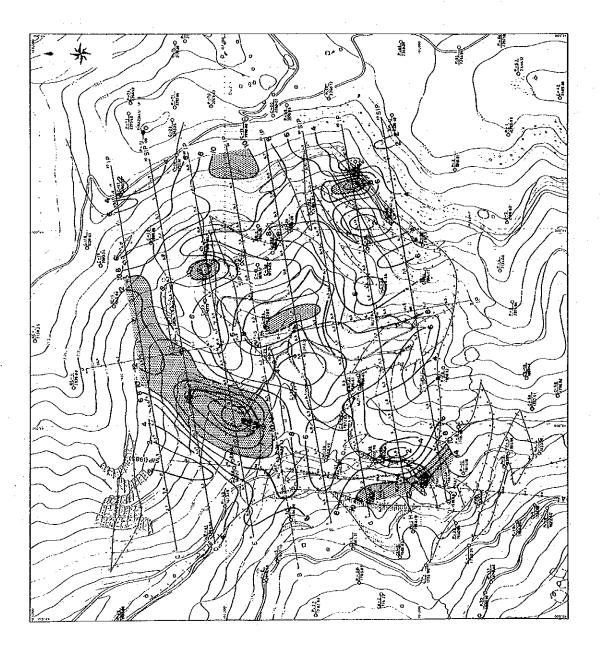
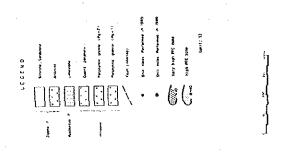


Fig. 21 Plan Map of PFE [0.125 - 1.0 Hz] (n = 1)



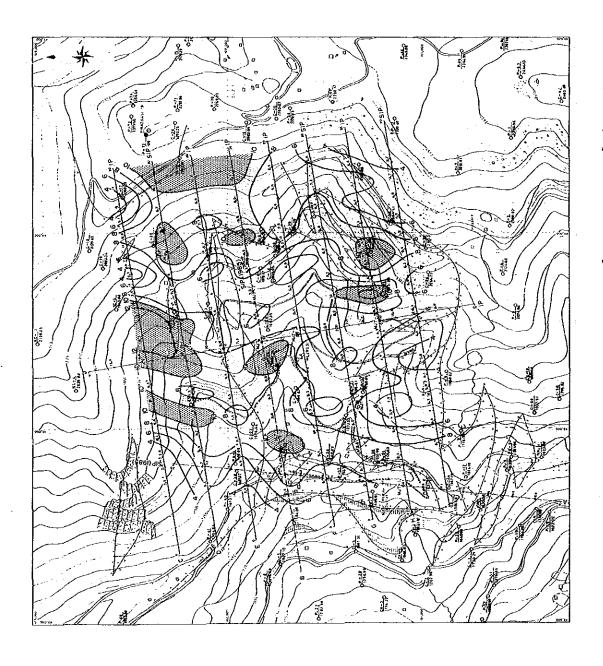


Fig. 22 Plan Map of PFE  $[0.125 - 1.0 \, \text{Hz}] \, (n = 3)$ 

N=5 plan (Fig.23): Areas of high PFE are similar to those of N=3 but zones of more than 10% are small in scale. They are at the central part of Lines C and D, in the vicinity of Nos.8-9 on Lines F and B, No.15 of Line B, No3 and No.16 of Line G, and in the vicinities of Nos.11 and 12 of Line H. They form a small U-shape.

## Section of PFE (Fig.24)

Continuity of anomalies of high PFE to depths is illustrated in the panel diagram.

The anomaly from the centre of Line C to the west of Line E originated from a source of anomaly at shallow depth. The anomaly extending towards the southeast shows a contour pattern derived from a source at a shallow depth. Anomalies of Nos.7-10 and 12-16 of Line B come from sources at medium to deep depths. The anomaly extending from the central part of the area to the south is also deemed to have been derived from a source at medium to deep depths.

Distribution characteristics gave three anomalies, namely, the horseshoe shape in the north, one extending from the central part to the south, and an anomaly along Maden River. Their origins are assumed to be situated at shallow depths except for the one in the centre of the area.

### (3) Pseudo-Section of Phase

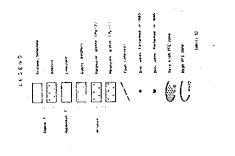
Sections were provided for each phase of 0.125, 0.375, 0.625, 1 and 3 Hz to investigate reliance of phase on frequencies. Phase sections after decoupling were also given.

## Line D (Fig. 25):

Sections of five frequencies shows similar patterns indicating that electromagnetic coupling was rare. Between No.5 and No.11, an area of 80 mrad shows a tendency to increase in accordance with an increase of frequency showing a little electromagnetic coupling. As observed in Figs.17-20, low values of AR less than 10 ohm-m are scarce and 86.4% of resistivities of rocks and cores exceeded 500 ohm-m. Consequently, it can be assumed that the resistivity of the area is large and that electromagnetic coupling phenomena are rare.

Phase sections after decoupling gave almost identical patterns at five frequencies. These patterns coincide with a phase pattern of fundamental frequency at 0.125 Hz indicating that electromagnetic coupling had been eliminated.

## Line G (Fig.26):



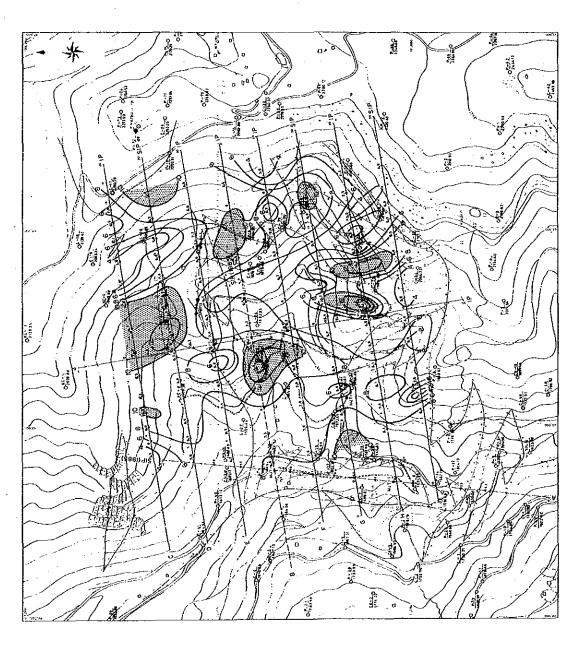


Fig. 23 Plan Map of PFE [0.125 - 1.0 Hz] (n = 5)

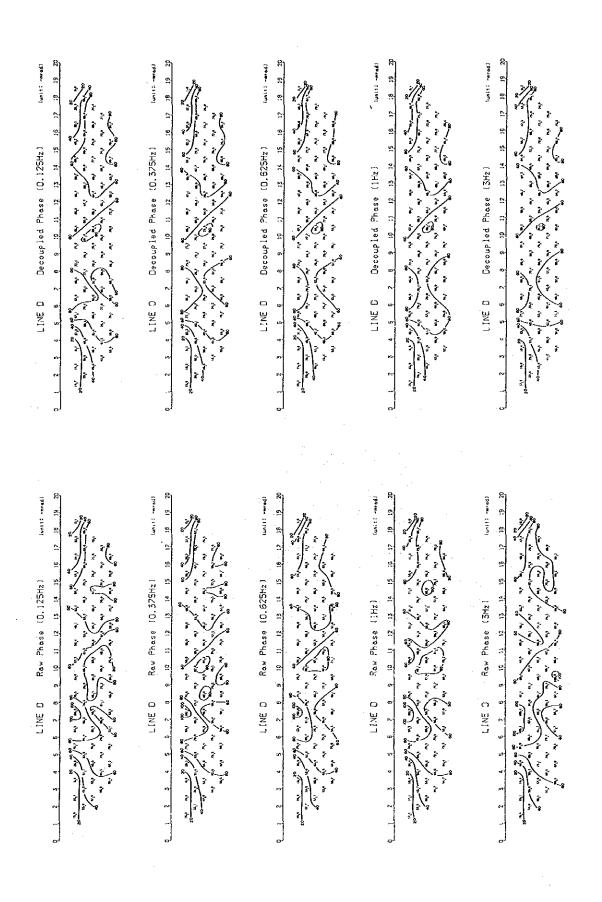
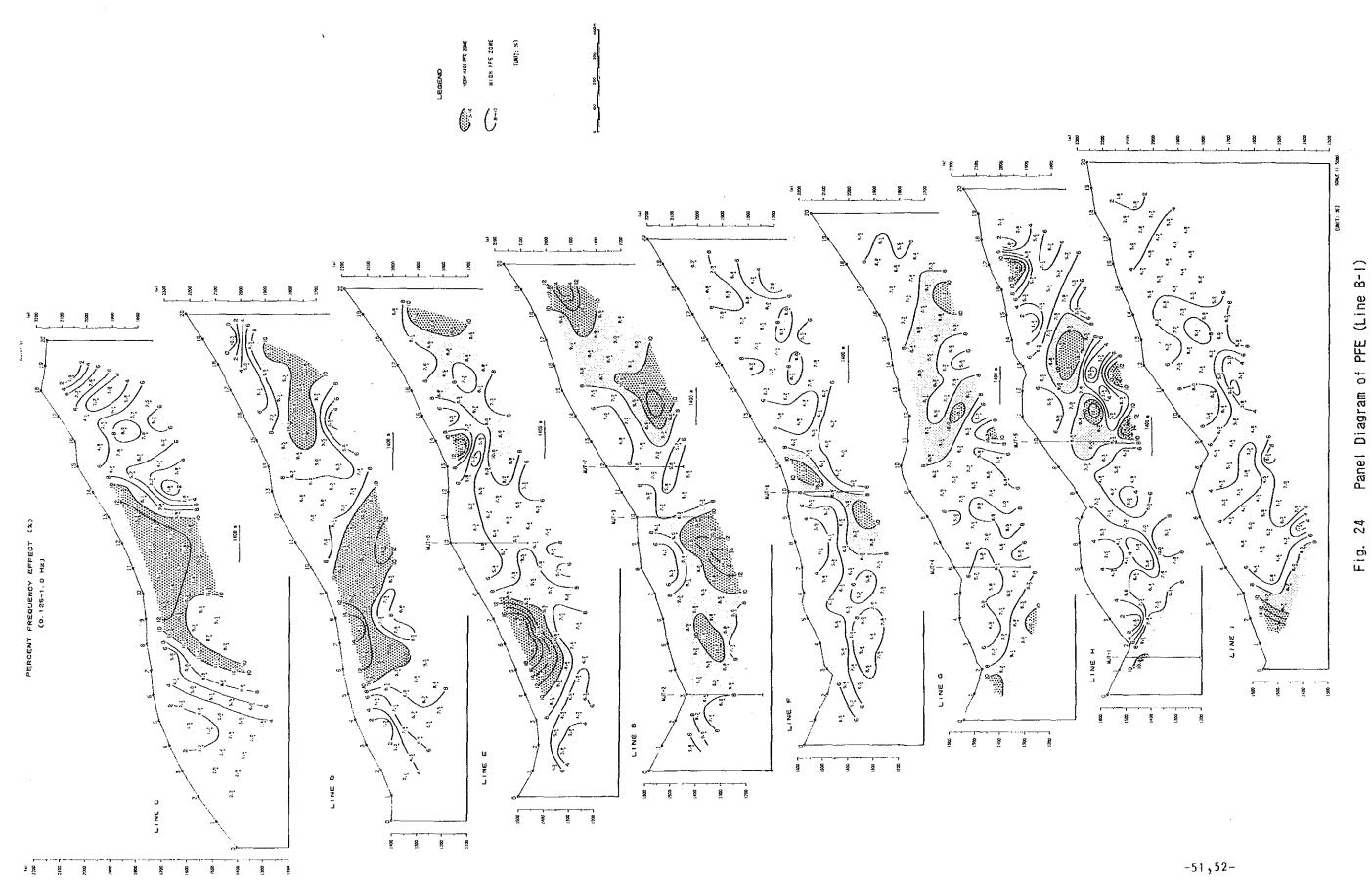


Fig. 25 Phase Difference at Five Frequencies (Line D)



-51,52-

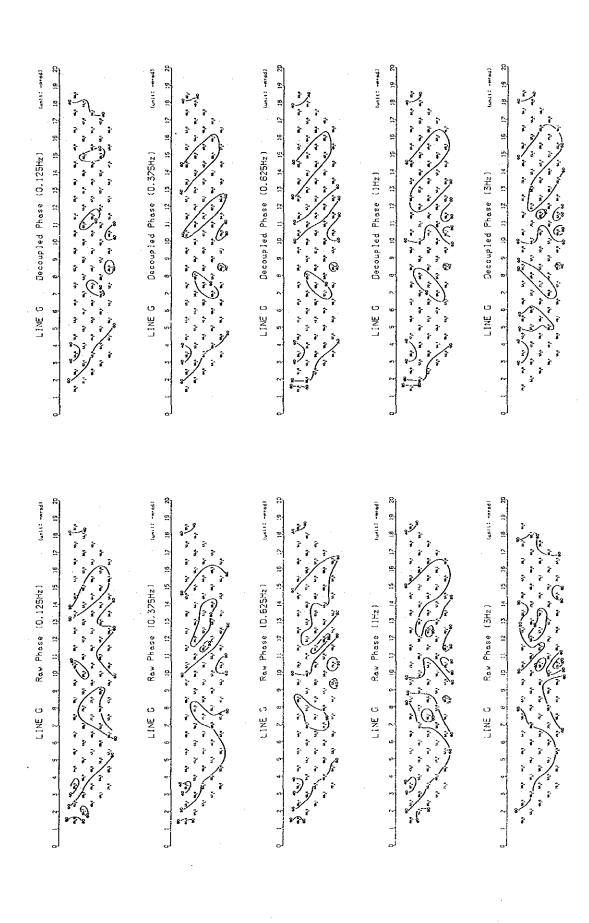


Fig. 26 Phase Difference at Five Frequencies (Line G)

In accordance with an increase of frequency, an area of 60 mrad becomes larger in the lower parts between No.6 and No.9 and between No.17 and No.19, showing a small effect of electromagnetic coupling at a frequency over 1 Hz. Phase sections on five frequencies after decoupling gave similar patterns with a section of fundamental frequecy (0.125 Hz) showing an effect of decoupling procedure.

# Line I(Fig.27):

Sections of five frequencies gave intimate contour patterns, as with the previous two lines, but a weak effect of electromagnetic coupling was noticed in accordance with an increase of frequency. Five sections after decoupling showed similar patterns as the one for the frequecy frequency (0.125 Hz), suggesting an effect of decoupling.

# (4) Phase, Magnitude and Cole-Cole Spectrum

Pseudosections of phase spectrum, magnitude spectrum and Cole-Cole diagram are provided. Various spectrum sections after decoupling are also presented.

## Line D (Fig. 28):

### (A) Phase spectrum

The predominant spectra have a form which is flat in the field up to 10 Hz and has a rising tendency beyond this frequency. In the lower part between No.5 and No.9, a flat spectrum was obtained over a whole area of frequencies. This feature also appears in the spectrum after decoupling, being convex between No.5 and No.9, or of a spectrum slanting down to the right-hand side (X-, Y-type). Unaccountable spectra were observed at depth of No.3 and No.18, where PFE stoppd at 1.2% and -1.3% respectively, being extremely low. Many spectra after decoupling are generally flat or of a dropping type.

### (B) Magnitude spectrum

Generally, spectra are of a dropping type with a small inclination. Larger inclination has been detected at depth between No.5 and No.9, as seen in the case of phase spectrum. Spectra after decoupling show larger inclinations at the same place. Similar spectra are observed between No.14 and No.15 at the level of N=1

## (C) Cole-Cole diagram

Many are C-type spectra slanting down. A B-type flat spectrum is detected in the same position where anomalous spectra of phase and magnitude have been located. Decoupling of Cole-Cole diagram generally gives upturn A-type spectra.

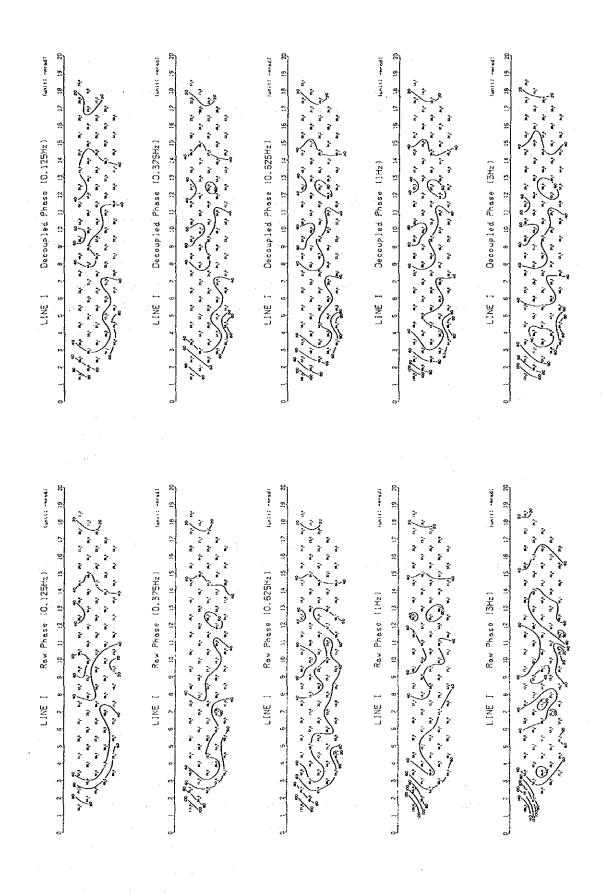


Fig. 27 Phase Difference at Five Frequencies (Line 1)

Spectrum anomalies are noticed between No.5 and No.9, and between No.14 and No.15, where PFE values stand at more than 12% and 8% respectively.

# Line G (Fig. 29):

## (A) Phase spectrum

Many spectra are of a monotonously ascending type, although noise increases beyond the depth of N=3. Three spectra which decline at a frequency beyond 8 Hz have been detected at the western end of the line. Many spectra are horizontal after decoupling, but downturn spectra are found at depth between No.2 and No.3, and upturn spectra are observed at depths at No.10 and No.15 and at moderate depth at No.11. An area where high values of PFE and phase have been detected between No.10 and No.16 shows an X-type spectrum in which the central part is convex.

#### (B) Magnitude spectrum

Spectra of downturn tendency with a uniform inclination are common. Different spectra have been detected after decoupling at the western end and at depth at No.10, where different spectra of up-and downturn spectra have been detected in the phase spectra.

# (C) Cole-Cole diagram

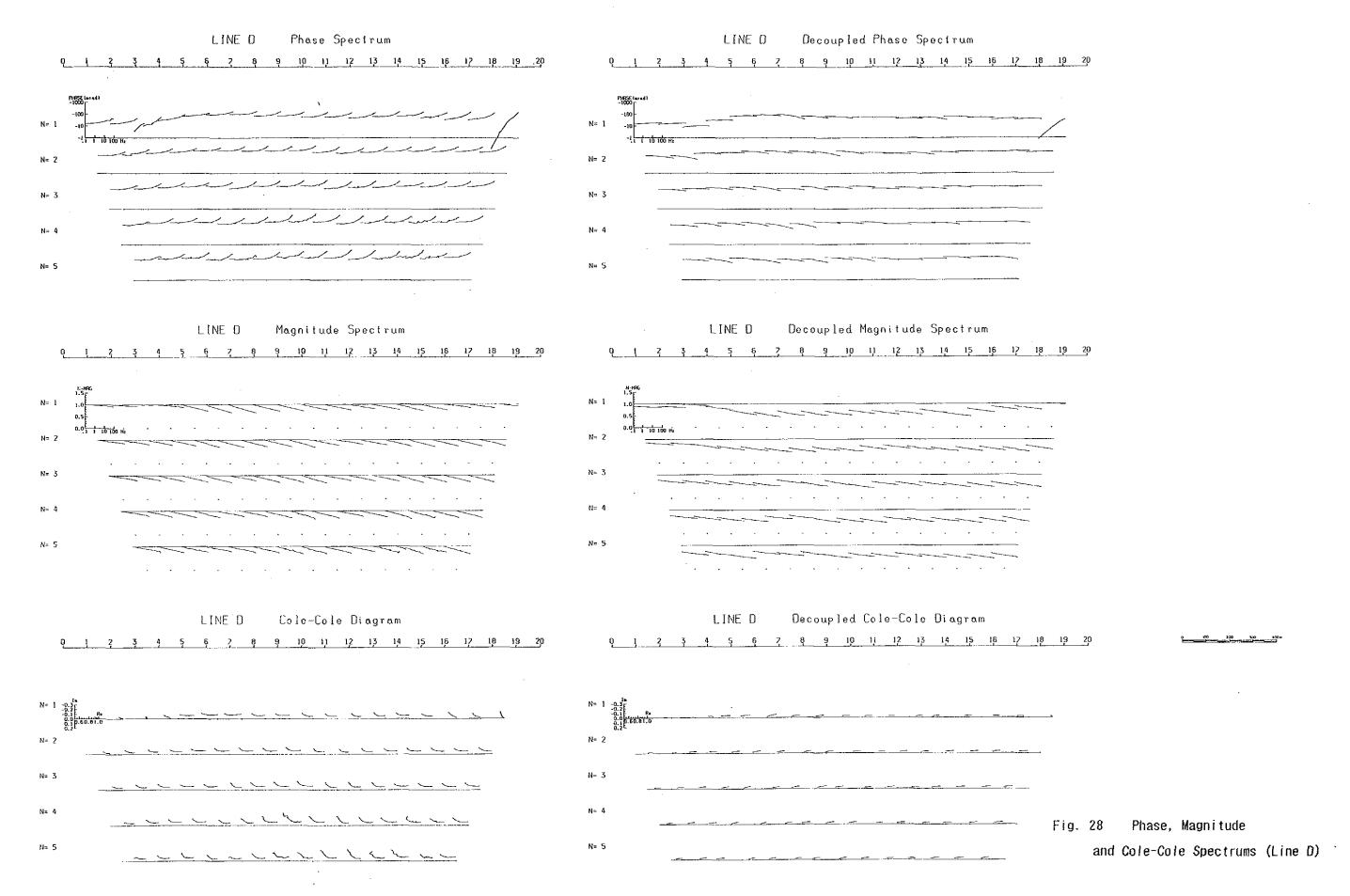
All spectra are of the downturn type, except at the western end where an upturn spectrum is found. After decoupling, many spectra are flat, but several spectra of upturn tendency are seen at the western end of the line.

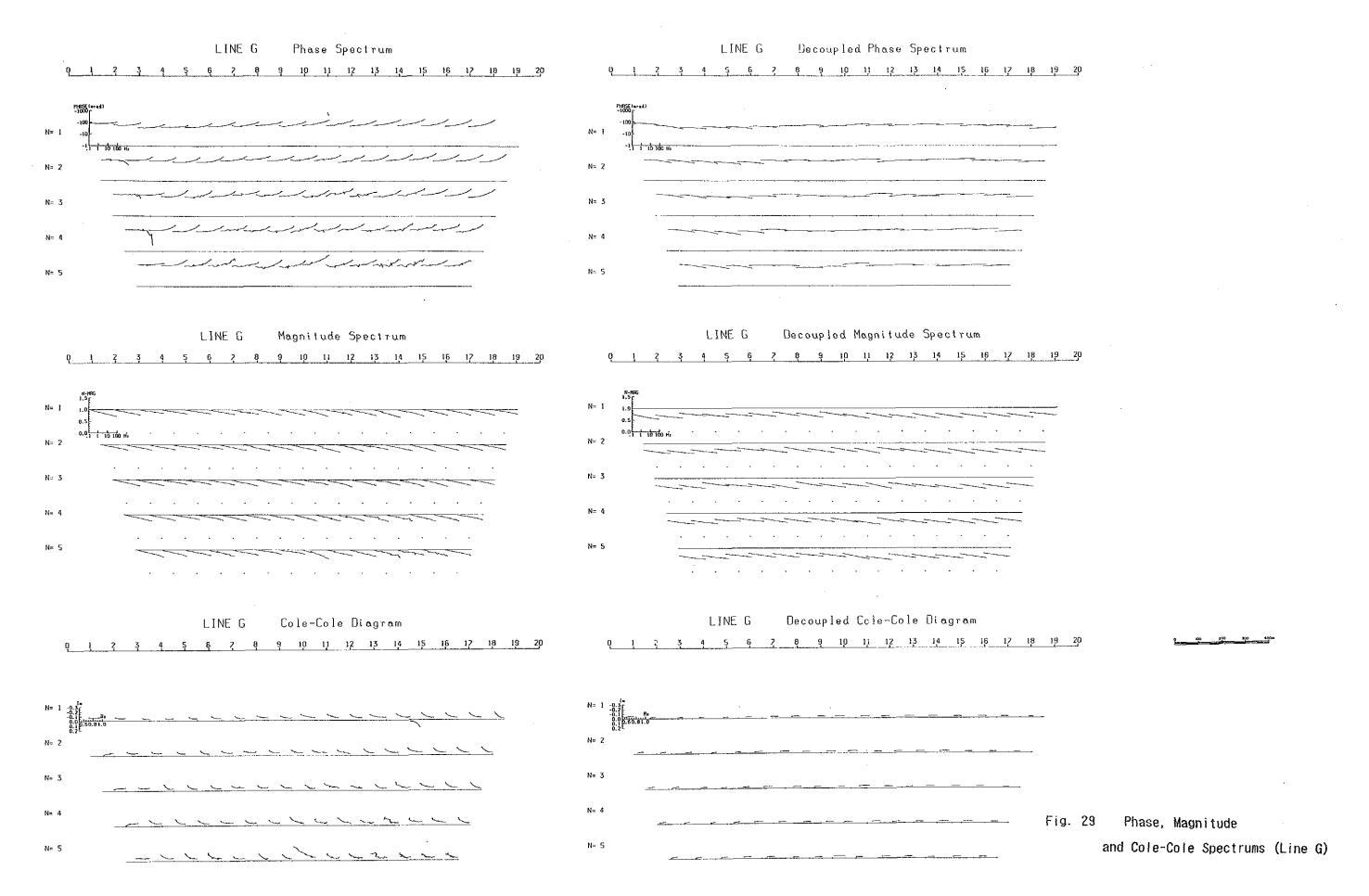
From the above, anomalies of spectra are pointed out at the western end of the line, at depths at Nos.10 and 15, and at moderate depth at No.11 where more than 10% PFE has been detected.

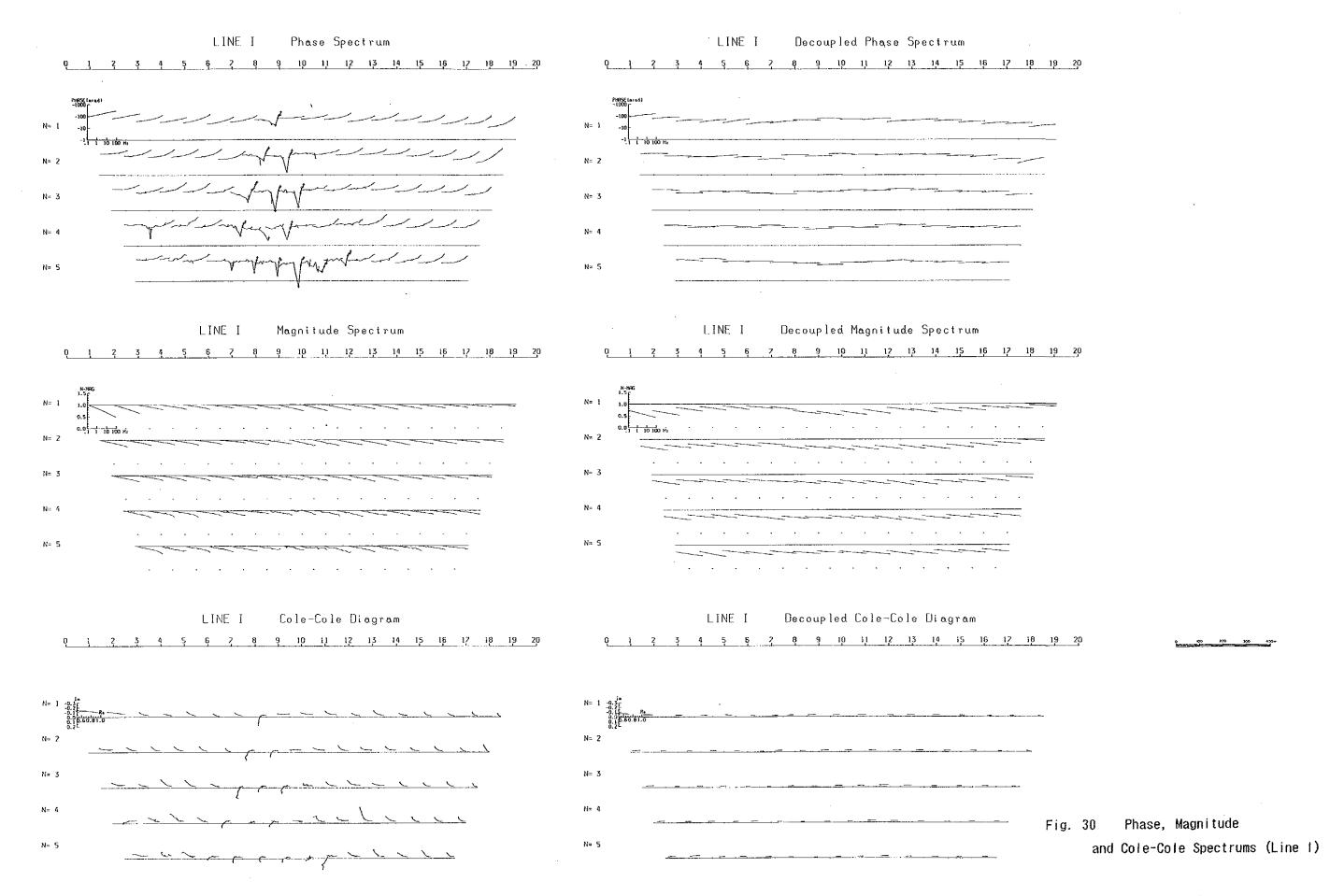
## Line I (Fig. 30):

#### (A) Phase spectrum

A negative phase has been detected at depth between No.6 and No.11. A high phase contrast was recorded at the western end of the line. Hasan River is situated between Nos.8 and 9 where a mineralization and alteration zone exists. Although a value of PFE is a little bit higher at around 6%, a zone of less than 50 ohm-m is distributed to give a large contrast against the AR of surrounding areas. After decoupling, upturn spectra are observed at both ends of the line, and convex or flat-lying spectra are observed in the middle of the line. The area of negative phase shows flat-lying spectra and a large contrast of phases.







# (B) Magnitude spectrum

No conspicuous characteristics have been recorded except downturn spectra with steep inclinations at the western end of the line. Magnitude after decoupling gives steep spectra west of the line and at shallow depth between No.8 and No.11.

#### (C) Cole-Cole diagram

Spectra are generally of a downturn trend except a central part of the line where a peculiar L-shape spectrum has been obtained in the section of negative phase. After decoupling, many spectra are of flat-lying B-type with a small number of C-type of downturn tendency in the vicinity of Nos.4 and 5. At the western end of the line, an upturn spectrum, which indicates mineralization, has been detected and is assumed to be related to a zone of pyrite impregnation in andesites of Maden river.

From the above, anomalies of spectra are pointed out at the western end of the line and at shallow depth between No.8 and No.11. The PFE of these areas exceed 8%.

# (5) Decoupled Percent Frequency Effect

Phase spectrum, magnitude spectrum and Cole-Cole diagram after decoupling have been discussed above. The PFE of D,G and I is explained in this section.

### Line D (Fig. 31, the top)

A contour pattern of PFE is similar with that of Fig.24, but a zone of 10% becomes larger between No.14 and No.18.

### Line G (Fig. 31, the middle)

The section has a similar pattern with that of Fig.24 but zones of more than 10% PFE at depths at No.3 and between No.15 and No.16 have disappearred.

# Line I (Fig. 31, the bottom)

Not much difference has been observed from before and after decoupling. A zone of high PFE at 10% occurs at the lower part of No.4.

Contour patterns of these three sections did not show apparent changes due to decoupling, indicating that the electromagnetic coupling did not give much influence. This phenomenon probably comes from high resistivities of rocks in the area.

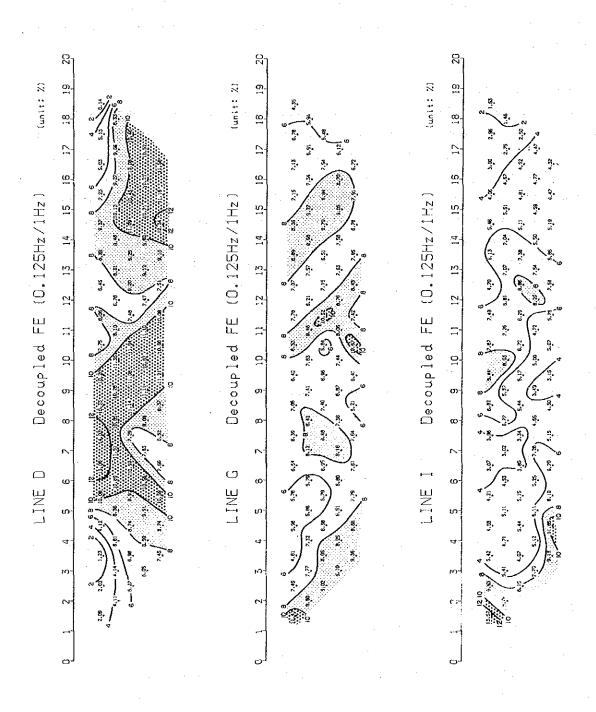


Fig. 31 Decoupled PFE (Line D, G and 1)

#### (6) Model Simulation for IP Anomaly

For a quantitative analysis, localities of anomaly sources and values of PFE and AR were investigated by model simulation. Centered at the drilling sites, the simulation was conducted on Lines B, E, F and J.

# Line B (Fig. 32-1):

High anomalies of PFE were delineated west of MJT-7 on the line. The area is mainly underlain by porphyritic granite (Pg1) except to the east where andesites occur. In the simulation, the porphyritic granite was coded as 8, the mineralized part as 3 and a leached zone near the surface as 1. The andesites were allocated Code 6. Simulation gave a distribution of AR and a pattern of PFE which is in accord with the field measurements, indicating that the given model is adequate. The mineralized zone (Code 3) is assumed to be the origin of the anomaly detected in the lower section between No.7 and No.10.

#### Line E (Fig. 32-2):

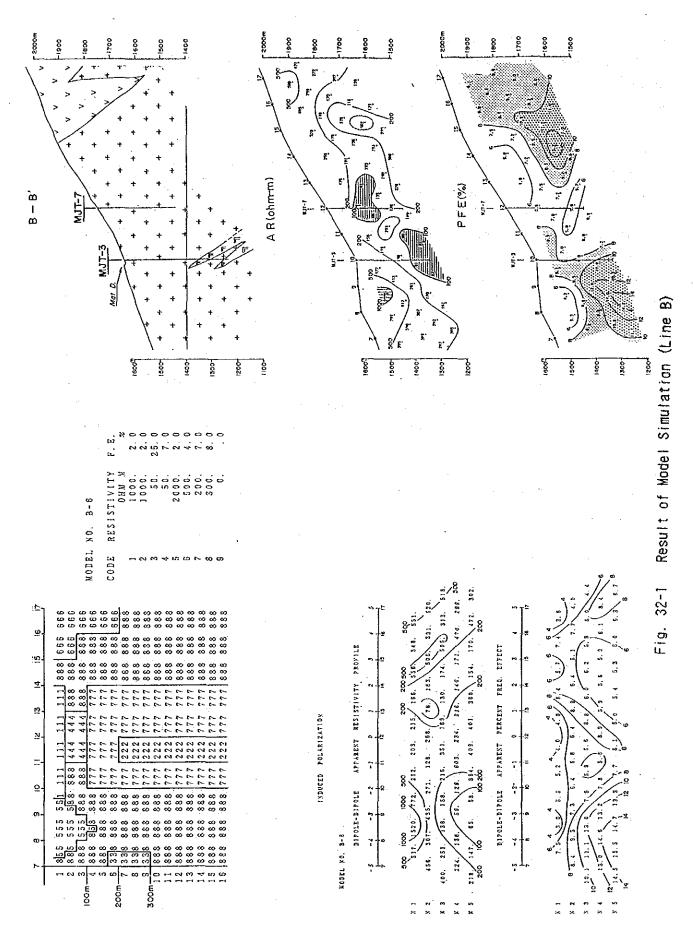
The most prominent anomaly has been delineated at depth between No.4 and No.8. The western half is overlain by andesites Coded 5, and the eastern half is underlain by the porphyritic granite (Pg1) Coded 5. A source of anomaly is considered to be at shallow depth and is given Code 1 and 2, and a leached zone is assigned Code 3.

Patterns of AR and PFE are generally similar with field measurements. A pantleg pattern is common, but the delineated pattern is hardly obtained by the simulation. Similar patterns obtained in the field are considered to be a result of the three dimensional aspect.

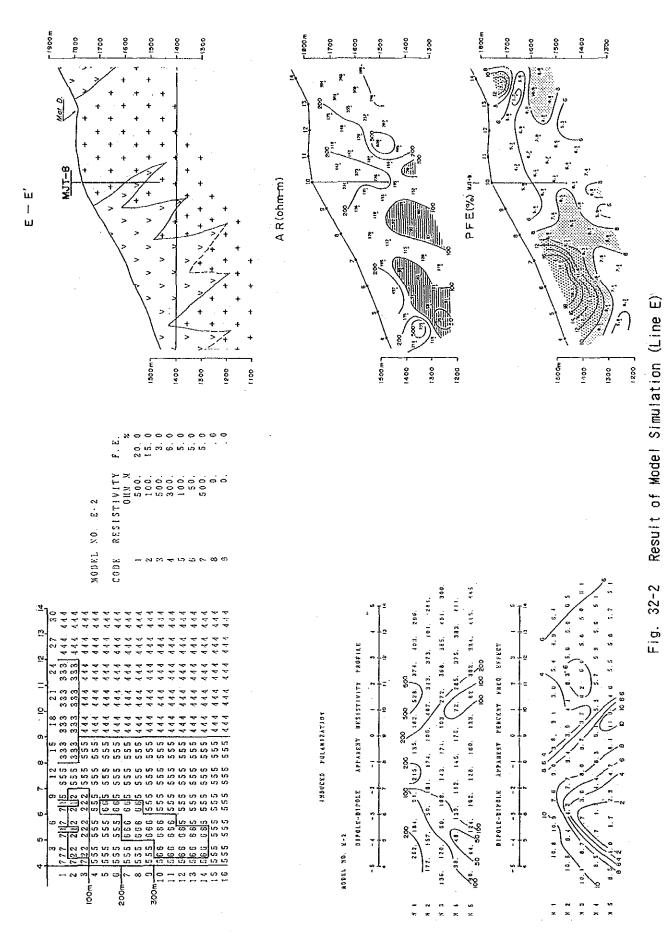
# Line F (Fig. 32-3):

An anomaly at a shallow depth is located east of MJT-6, and an anomaly at depth is in the west. The porphyritic granite (Pg1) (Codes 4 and 5) is distributed along the line. An intrusive rock of porphyritic granite (Pg2) was penetrated by drilling. Highly resistive areas were given Codes 7 and 9, and strongly pyritized sections were allotted with Codes 3 and 8. The penetrated intrusive of Code 2 is considered to have a similar AR as the porphyritic granite (Pg1), and a value of PFE was estimated at a level of 2% due to a lack of mineralization.

After the simulation, AR and PFE were in accordance with the field measurements. Areas Coded 3 and 8 of with high PFE probably correspond to sections of strong pyritization. A zone of high AR with low PFE (Codes 7 and 9) at shallow depth may be the leached zone.



-66~



-67-

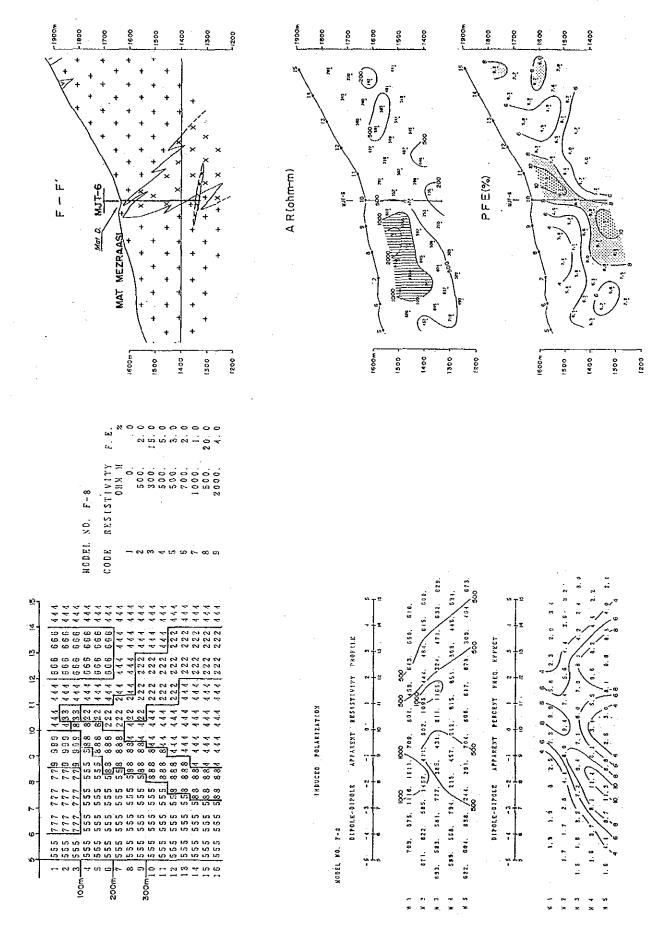


Fig. 32-3 Result of Model Simulation (Line F)

Fig. 33 Pseudo-Section of Apparent Resistivity and PFE (Line J)

#### Line J (Fig. 33):

The line traverses the central part in a N-S direction. After consultation with MTA, the standard IP method was conducted to delineate a distribution of high anomalies of PFE.

A zone of high AR (over 1,000 ohm-m) has not been observed; zones of AR less than 100 ohm-m were detected in the following three places:

- 1 between No.4 and No.7 on andesites
- ② between No.9 and No.11 on the porphyritic granite (Pg1)
- 3 between No.14 and No.18 on the same Pgl

These zones of low AR are situated in the lower parts of topographic depressions. Topographic effects have been eliminated by the terrain correction and these are related with groundwater and alteration zones. The zone of ③ corresponds with a zone of mineralization and alteration detected at Hasan River.

The values of PFE exceeded 6% over the whole area and areas of more than 10% have been detected in four places:

- (1) northern part of No.1 in andesites
- ② between No.2 and No.6 in andesites
  - (3) between No.7 and No.12 in an area of andesite and porphyritic granite(Pg1)
  - (4) between No.14 and No.17 in the porphyritic granite (Pg1)

From patterns of PFE, (1), (2) and (4) are of origins continuous from near surface to depth, and (3) may be related to an origin at shallow depth.

The simulation was conducted over an area between No.3 and No.13 ( See Andesites at the northern part of MJT-8 (Code 5) and the porphyritic granites (Pg1) (Code 6) in the southern part have been taken into consideration. Strong mineralization is assumed in the lower part of No.3 (Code 1) and between No.8 and No.10 (Code 2). Dacites intersected by MJT-6 The result of the simulation is harmonic with PFE in the were given Code 3. field, but some differences in AR have been noticed between calculated values Calculated resistivities are higher than the field and field measurements. measurements for areas detected at depths between No.4 and No.7 and between No.9 and No.12. However, delineated anomalies give enough of a basis to estimate a model of PFE distribution.

# (7) Discussion and Interpretation Map

The survey is comprised of the measurements of SIP over three lines totalling 6 km and IP over six lines totalling 12 km with a line interval of 200 m to delineate the extension of the mineralized zone.

The results are summarized as follows.

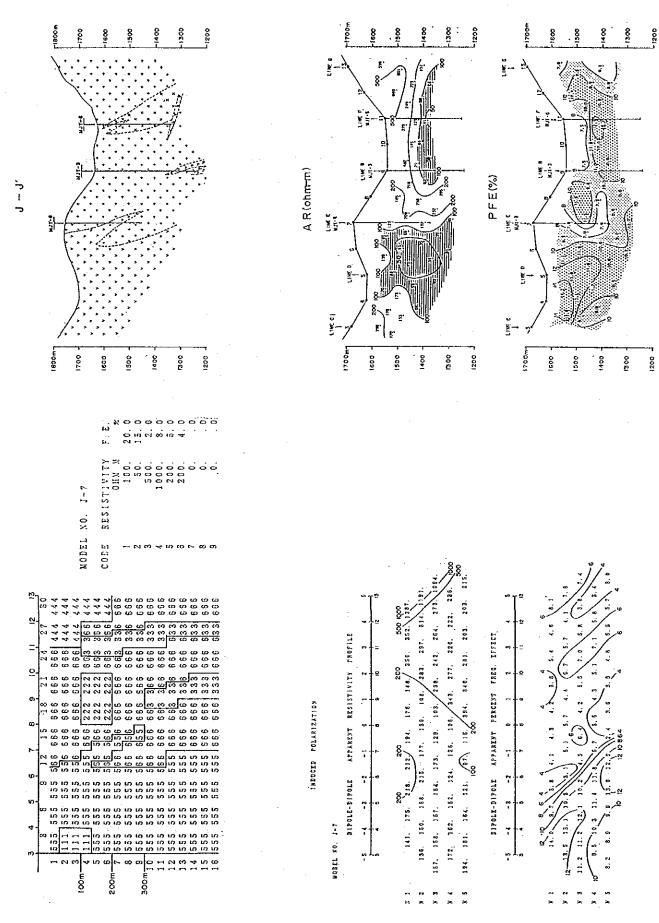


Fig. 34 Result of Model Simulation (Line J)

#### 1) AR

Prevailing AR are of the order of 500 ohm-m and zones of AR less than 100 ohm-m have been detected mainly in the vicinities of Maden River in the west and Hasan River in the south. These zones are situated in areas of andesites and the porphyritic granite (Pg1), and are probably related with alteration zones and ground-water. Zones of high AR are located between No.6 and No.9 and between No.13 and No.17 of Line C in the north and east of Lines H and I in the south-eastern part of the area where andesites and porphyritic granites underlie.

#### 2) PFE

The PFE stands as an indicator of the extension of the mineralized zone and more than 90 % of the PFEs exceeded a level of 4 %. The laboratory investigation of surface rock samples and cores conducted in the previous year showed that the PFE of fresh rocks or weakly mineralized samples did not exceed 1.5 %. A zone of 4 % PFE can be deemed anomalous, suggesting that the whole area belongs to a mineralized zone. A zone of more than 8 % PFE gives a horseshoe shape distribution in the northern half of the area, and another zone of high PFE has been detected in the central part of the area extending southward (Figs. 21 and 24).

In general, ore deposits of porphyry copper type have a zonal distribution of alteration zones consisting of, from outside to inside, propylitic, phyllic and potassic zones. A pyrite shell is found on the outer side of the phyllic alteration zone and a bonanza of copper and molybdenum is located inside the pyrite shell. The horseshoe shape zone of more than 8 % PFE at N=1 can be correlated with the pyrite shell and roughly coincides with a phyllic zone delineated by the geological survey of last year, and is assumed to extend to the poximity of Line D in the north-eastern area as indicated by this investigation. A geochemical anomaly of more than 200 ppm Cu lies in the zone of high PFE and the anomaly of more than 75 ppm of Mo is situated inside this anomalous zone.

#### 3) Rock properties

Rock properties of 25 drilling cores from this year are given in Table 4-1 with the results of 19 samples of the previous year. The relation between phase and metal content of Cu and Mo is shown in Table 4-2. Rock properties of each phase type are given in Table 4-3 with the minimum (Min.), the maximum (Max.) and the average (Ave.) values of metal content.

According to Table 4-2, a specific type of phase spectrum does not correspond

Table 4-1 Relationship between SIP Response and Mineralized Zone

No. (m)	artz vein  byrite along fissures  ite  ide  idiss.pyrite  idiss.pyrite  te,sericite-chlorite  idiss.pyrite  te along fissures
Drilling No.   HJT - 1	te srtz vein syrite along fissures te
MJT - 1	te srtz vein syrite along fissures te
21   52.10   Alternated andesite   -2.2   0.50   162   D   0.06   0.001   Sericite   C   22   99.80   Alternated andesite   -4.3   -0.19   126   D   0.09   0.009   Diss.pyrite   C   23   139.90   Porphyrtic granite(pg1)   21.8   4.56   219   D   0.08   0.004   Pyrite-quark   C   24   150.80   Alternated andesite   561.6   134.17   449   Y   0.10   0.001   Epidote, proceed   C   25   184.50   Porphyrtic granite(pg1)   208.2   38.71   1.795   X   0.09   0.019   Diss.pyrite   C   200.10   Basaltic andesite   3.5   0.45   4.757   A   0.06   0.001   Propyliza   C   250.90   Andesite   S.8   1.43   1.299   A   0.03   0.00   Filmy pyrite   C   27   250.90   Andesite   117.8   18.20   547   X   0.04   Propyliza   C   29   297.90   Basaltic andesite   376.2   38.31   1.361   X   0.01   Propyliza   C   29   297.90   Basaltic andesite   6.7   1.49   824   D   0.40   0.010   Propyliza   C   23   251.70   Alternated andesite   6.7   1.49   824   D   0.40   0.010   Propyliza   C   23   251.70   Alternated andesite   221.2   37.22   2.035   Y   0.12   0.003   Diss.pyrite   C   250.10   Alternated andesite   221.2   37.22   2.035   Y   0.12   0.004   Diss.pyrite   C   250.10   Alternated andesite   21.2   3.30   10.068   A   0.19   0.008   Diss.pyrite   C   250.10   Alternated andesite   22.2   3.30   10.068   A   0.19   0.008   Diss.pyrite   C   250.10   Alternated andesite   22.2   3.30   10.068   A   0.19   0.004   Diss.pyrite   C   250.10   Alternated andesite   22.2   3.30   10.068   A   0.19   0.004   Diss.pyrite   C   250.10   Alternated andesite   22.2   3.30   10.068   A   0.13   0.004   Diss.pyrite   C   250.10   Alternated andesite   22.2   3.30   10.068   A   0.13   0.004   Diss.pyrite   C   250.10   Diss.pyrite   C   250	te srtz vein syrite along fissures te
22   99.80   Alternated andesite   -4.3   -0.19   126   D   0.09   0.009   Diss.pyrite   23   139.90   Porphyrtic granite(pg1)   21.8   4.56   219   D   0.08   0.004   Pyrite-quarter   24   150.80   Alternated andesite   561.6   134.17   449   Y   0.10   0.001   Epidote, pg   25   184.50   Porphyrtic granite(pg1)   208.2   38.71   1.795   X   0.09   0.019   Diss.pyrite   26   200.10   Basaltic andesite   3.5   0.45   4.757   A   0.06   0.001   Propyliza   27   250.90   Andesite   8.8   1.43   1.299   A   0.03   0.00   Filmy pyrite   28   274.30   Basaltic andesite   117.8   18.20   547   X   0.04   Propyliza   29   297.90   Basaltic andesite   376.2   38.31   1.361   X   0.01   Propyliza   29   297.90   Basaltic andesite   376.2   38.31   1.361   X   0.01   Propyliza   31   44.70   Porphyrtic granite(pg1)   7.8   1.60   294   D   0.15   0.003   Diss.pyrite   32   51.70   Alternated andesite   6.7   1.49   824   D   0.40   0.010   Propyliza   33   154.30   Alternated andesite   541.2   144.57   550   X   0.12   0.003   Diss.pyrite   35   250.10   Alternated andesite   221.2   37.22   2.035   Y   0.12   0.004   Diss.pyrite   36   299.30   Basaltic andesite   21.2   3.30   10.068   A   0.19   0.008   Diss.pyrite   36   299.30   Basaltic andesite   32.1   4.91   4.303   B   0.13   0.004   Diss.pyrite   37   38   39.30   Basaltic andesite   32.1   4.91   4.303   B   0.13   0.004   Diss.pyrite   37   38   39.30   Basaltic andesite   32.1   4.91   4.303   B   0.13   0.004   Diss.pyrite   37   38   39.30   Basaltic andesite   32.1   4.91   4.303   B   0.13   0.004   Diss.pyrite   38   38   38   38   38   38   38   3	te srtz vein syrite along fissures te
23   139.90   Porphyrtic granite(ps1)   21.8   4.56   219   D   0.08   0.004   Pyrite-que   24   150.80   Alternated andesite   561.6   134.17   449   Y   0.10   0.001   Epidote, principal   25   184.50   Porphyrtic granite(ps1)   208.2   38.71   1.795   X   0.09   0.019   Diss.pyrite   26   200.10   Basaltic andesite   3.5   0.45   4.757   A   0.06   0.001   Propyliza   27   250.90   Andesite   8.8   1.43   1.299   A   0.03   0.00   Filmy pyrite   28   274.30   Basaltic andesite   117.8   18.20   547   X   0.04   Propyliza   29   297.90   Basaltic andesite   376.2   38.31   1.361   X   0.01   Propyliza   Drilling No.   NJT - 2   31   44.70   Porphyrtic granite(ps1)   7.8   1.60   294   D   0.15   0.003   Diss.pyrite   32   51.70   Alternated andesite   6.7   1.49   824   D   0.40   0.010   Propyliza   33   154.30   Alternated andesite   541.2   144.57   550   X   0.12   0.003   Diss.pyrite   35   250.10   Alternated andesite   221.2   37.22   2.035   Y   0.12   0.004   Diss.pyrite   36   299.30   Basaltic andesite   21.2   3.30   10.068   A   0.19   0.008   Diss.pyrite   36   299.30   Basaltic andesite   32.1   4.91   4.303   B   0.13   0.004   Diss.pyrite   37   38   39.30   Basaltic andesite   32.1   4.91   4.303   B   0.13   0.004   Diss.pyrite   37   38   39.30   Basaltic andesite   32.1   4.91   4.303   B   0.13   0.004   Diss.pyrite   37   38   39.30   Basaltic andesite   32.1   4.91   4.303   B   0.13   0.004   Diss.pyrite   38   39.30   Basaltic andesite   32.1   4.91   4.303   B   0.13   0.004   Diss.pyrite   39.30   Diss.pyrite   30.30	artz vein  byrite along fissures  ite  ide  idiss.pyrite  idiss.pyrite  te,sericite-chlorite  idiss.pyrite  te along fissures
24         150.80         Alternated andesite         561.6         134.17         449         Y         0.10         0.001         Epidote, i           25         184.50         Porphyrtic granite(pg1)         208.2         38.71         1.795         X         0.09         0.019         Diss.pyrit           26         200.10         Basaltic andesite         3.5         0.45         4.757         A         0.06         0.001         Propyliza           27         250.90         Andesite         8.8         1.43         1.299         A         0.03         0.00         Filmy pyr           28         274.30         Basaltic andesite         117.8         18.20         547         X         0.04         Propyliza           29         297.90         Basaltic andesite         376.2         38.31         1.361         X         0.01         Propyliza           Drilling No.           NJT - 2         31         44.70         Porphyrtic granite(pg1)         7.8         1.60         294         D         0.15         0.003         Diss.pyrit           32         51.70         Alternated andesite         541.2         144.57         550         X         0.12	oyrite along fissures  Le  Le  Lodiss.pyrite  Le,sericite-chlorite  Lodiss.pyrite  Le along fissures
24       150.80       Alternated andesite       561.6       134.17       449       Y       0.10       0.001       Epidote, respectively         25       184.50       Porphyrtic granite(pg1)       208.2       38.71       1.795       X       0.09       0.019       Diss.pyrit         26       200.10       Basaltic andesite       3.5       0.45       4.757       A       0.06       0.001       Propyliza         27       250.90       Andesite       8.8       1.43       1.299       A       0.03       0.00       Filmy pyr         28       274.30       Basaltic andesite       117.8       18.20       547       X       0.04       Propyliza         29       297.90       Basaltic andesite       376.2       38.31       1.361       X       0.01       Propyliza         Drilling No. NJT - 2       81       44.70       Porphyrtic granite(pg1)       7.8       1.60       294       D       0.15       0.003       Diss.pyrit         32       51.70       Alternated andesite       541.2       144.57       550       X       0.12       0.003       Diss.pyrit         34       200.00       Alternated andesite       221.2       37.22	oyrite along fissures  Le  Le  Lodiss.pyrite  Le,sericite-chlorite  Lodiss.pyrite  Le along fissures
25	tediss.pyritediss.pyritediss.pyrite te.sericite-chloritediss.pyrite te along fissures
26   200.10   Basaltic andesite   3.5   0.45   4.757   A   0.06   0.001   Propyliza	itediss.pyritediss.pyritediss.pyrite te.sericite-chloritediss.pyrite te along fissures
27   250.90   Andesite   8.8   1.43   1.299   A   0.03   0.00   Filmy pyrical	tediss.pyritediss.pyrite te.sericite-chloritediss.pyrite te along fissures
28       274.30       Basaltic andesite       117.8       18.20       547       X       0.04       Propyliza         29       297.90       Basaltic andesite       376.2       38.31       1.361       X       0.01       Propyliza         Drilling No. NJT - 2         31       44.70       Porphyrtic granite(pg1)       7.8       1.60       294       D       0.15       0.003       Diss.pyrit         32       51.70       Alternated andesite       6.7       1.49       824       D       0.40       0.010       Propyliza         33       154.30       Alternated andesite       541.2       144.57       550       X       0.12       0.003       Diss.pyrit         34       200.00       Alternated andesite       221.2       37.22       2.035       Y       0.12       0.004       Diss.pyrit         35       250.10       Alternated andesite       21.2       3.30       10.068       A       0.19       0.008       Diss.pyrit         36       299.30       Basaltic andesite       32.1       4.91       4.303       B       0.13       0.004       Diss.pyrit         Brilling No.       MJT - 3       Drilling No.       A	.diss.pyrite .diss.pyrite te.sericite-chlorite .diss.pyrite te along fissures
29   297.90   Basaltic andesite   376.2   38.31   1.361   X   0.01   Propyliza	.,diss.pyrite te,sericite-chlorite .,diss.pyrite te along fissures
Drilling No.   NJT - 2     31	Le sericite-chlorite . diss.pyrite te along fissures
NJT - 2     31   44.70     Porphyrtic granite(pg1)   7.8   1.60   294   D   0.15   0.003   Diss.pyrit   32   51.70   Alternated andesite   6.7   1.49   824   D   0.40   0.010   Propyliza   33   154.30   Alternated andesite   541.2   144.57   550   X   0.12   0.003   Diss.pyrit   34   200.00   Alternated andesite   221.2   37.22   2.035   Y   0.12   0.004   Diss.pyrit   35   250.10   Alternated andesite   21.2   3.30   10.068   A   0.19   0.008   Diss.pyrit   36   299.30   Basaltic andesite   32.1   4.91   4.303   B   0.13   0.004   Diss.pyrit   Drilling No.   MJT - 3	. diss.pyrite te along fissures
31   44.70   Porphyrtic granite(pg1)   7.8   1.60   294   D   0.15   0.003   Diss.pyrit	. diss.pyrite te along fissures
32   51.70   Alternated andesite   6.7   1.49   824   D   0.40   0.010   Propyliza   33   154.30   Alternated andesite   541.2   144.57   550   X   0.12   0.003   Diss.pyrit   34   200.00   Alternated andesite   221.2   37.22   2.035   Y   0.12   0.004   Diss.pyrit   35   250.10   Alternated andesite   21.2   3.30   10.068   A   0.19   0.008   Diss.pyrit   36   299.30   Basaltic andesite   32.1   4.91   4.303   B   0.13   0.004   Diss.pyrit   Drilling No.   MJT - 3   Alternated andesite   32.1   4.91   4.303   B   0.13   0.004   Diss.pyrit   Drilling No.   MJT - 3   Alternated andesite   32.1   4.91   4.303   B   0.13   0.004   Diss.pyrit   Drilling No.   MJT - 3   Alternated andesite   32.1   4.91   4.303   B   0.13   0.004   Diss.pyrit   Drilling No.   MJT - 3   Alternated andesite   32.1   4.91   4.303   B   0.13   0.004   Diss.pyrit   Drilling No.   MJT - 3   Alternated andesite   32.1   4.91   4.303   B   0.13   0.004   Diss.pyrit   Drilling No.   MJT - 3   Alternated andesite   32.1   4.91   4.303   B   0.13   0.004   Diss.pyrit   Drilling No.   MJT - 3   Alternated andesite   32.1   4.91   4.303   B   0.13   0.004   Diss.pyrit   Drilling No.   MJT - 3   Alternated andesite   32.1   4.91   4.303   B   0.13   0.004   Diss.pyrit   Drilling No.   MJT - 3   Drilling No.   Alternated andesite   32.1   4.91   4.303   B   0.13   0.004   Diss.pyrit   Drilling No.   Alternated andesite   32.1   4.91   4.303   B   0.13   0.004   Diss.pyrit   Drilling No.   Alternated andesite   32.1   4.91   4.303   B   0.13   0.004   Diss.pyrit   Drilling No.   Alternated andesite   32.1   4.91   4.303   B   0.13   0.004   Diss.pyrit   Drilling No.   Alternated andesite   32.1   4.91   4.303   Drilling No.   Drilling No	. diss.pyrite te along fissures
33         154:30         Alternated andesite         541.2         144.57         550         X         0.12         0.003         Diss.pyrid           34         200.00         Alternated andesite         221.2         37.22         2.035         Y         0.12         0.004         Diss.pyrid           35         250.10         Alternated andesite         21.2         3.30         10.068         A         0.19         0.008         Diss.pyrid           36         299.30         Basaltic andesite         32.1         4.91         4.303         B         0.13         0.004         Diss.pyrid           Brilling No.         MJT - 3         Alternated andesite         32.1         4.91         4.303         B         0.13         0.004         Diss.pyrid	te along fissures
34         200.00         Alternated andesite         221.2         37.22         2.035         Y         0.12         0.004         Diss.pyrit           35         250.10         Alternated andesite         21.2         3.30         10.068         A         0.19         0.008         Diss.pyrit           36         299.30         Basaltic andesite         32.1         4.91         4.303         B         0.13         0.004         Diss.pyrit           Brilling No.         MJT - 3         Alternated andesite         32.1         4.91         4.303         B         0.13         0.004         Diss.pyrit	
35   250.10   Alternated andesite   21.2   3.30   10.068   A   0.19   0.008   Diss.pyrit   36   299.30   Basaltic andesite   32.1   4.91   4.303   B   0.13   0.004   Diss.pyrit   Drilling No.   MJT - 3	
36   299.30   Bassitic andesite   32.1   4.91   4.303   B   0.13   0.004   Diss.pyrium	
Drilling No. MJT - 3	te along fissures
MJT - 3	te along fissures
41 151.05 Porphyrtic granite(pg1) 36.7   5.53   849   A   0.21   0.006   Sil. serio	ite,molybdenite-qz
42 199.20 Porphyrtic granite(ps1) 32.4 4.69 3.754 B 0.15 0.011 Sil. serie	cite.diss.pyrite
	odenite diss.pyrite
	ite diss.pyrite
Drilling No.	- Caroop 1110
HJT - 4	•
	te, sericite-biotite
	ite, diss.pyrite, biotite-chlorite
7 7 7 7 7 7 7 7 7 7 7 7 7 7 7 7 7 7 7 7	
	piotite-chlorite
	te, biotite-sericite
	chlorite-biotite
6 250.0 Porphyrtic granite(pg1) 28.7 3.7 852 Y 0.04 0.005 Biotite-se	
7 300.0 Porphyrtic granite(pg1) 19.5 2.4 1.100 Y.(E) 0.04 0.010 Sericite-	piotite-chlorite
Drilling No.	
NJT - 5	
8 49.0 Porphyrtic granite(pg1) 63.0 9.3 3,560 A 0.05 0.000 Chalcocite	e, Chalcopyrite, sericite
9 99.0 Porphyrtic granite(pg1) 41.2 6.6 2,160 D 0.03 0.000 Sericite	
10 195.0 Basaltic andesite 114.0 20.2 5.690 D.(B) 0.06 0.000	
Drilling No.	
HJT - 6	
11   13.3   Porphyrtic granite(pg1)   27.9   3.8   1,440   B   0.35   0.024   Sericite	
13 100.05 Porphyrtic granite(ps1) 18.7 2.5 892 0.(E) 0.31 0.010 Scricite-o	CHIOLIES
14 150.0 Porphyrtic granite(pg2) 17.5 2.6 3.720 B 0.01 0.000	
15 198.8 Porphyrtic granite(pg2) 40.1 6.0 1,580 B 0.06 0.000	
16 250.0 Porphyrtic granite(pg1) 27.6 5.2 889 D 0.19 0.014	
17   301.0   Porphyrtic granite(pg2)  20.5   2.4   405   E   0.09   0.002   Silicified	
Drilling No.	
MJT - 7	
18 16.0 Porphyrtic granite(pg1) 16.4 2.7 7,300 D 0.16 0.013 Sericite,	gurtz vein
19 55.0 Porphyrtic granite(pg1) 54.5 10.2 2,530 D 0.05 0.020 Sericite	
20 255.0 Porphyrtic granite(pg1) 47.9 8.3 1.600 D 0.15 0.018 Sericite.	qurtz vein
21 300.0 Porphyrtic granite(pg1) 62.7 8.7 4,200 Y 0.10 0.006 Sericite-s	
Drilling No.	
HJT - 8	
22 36.5 Porphyrtic granite(pg1) 24.0 3.7 8,700 A 0.03 0.001 Sericite	
23 46.6 Qurtz vein 13.8 2.2 3,120 A 0.04 0.000	
24 146.35 Porphyrtic granite(pg1) 20.3 2.7 911 E 0.08 0.015 Sericite.	
25 190.1 Porphyrtic granite(pg1) 31.2 3.9 195 E 0.14 0.006 Sericite,	qurtz vein

1985: MJT-1,-2,-3 1986: MJT-4,-5,-6,-7,-8

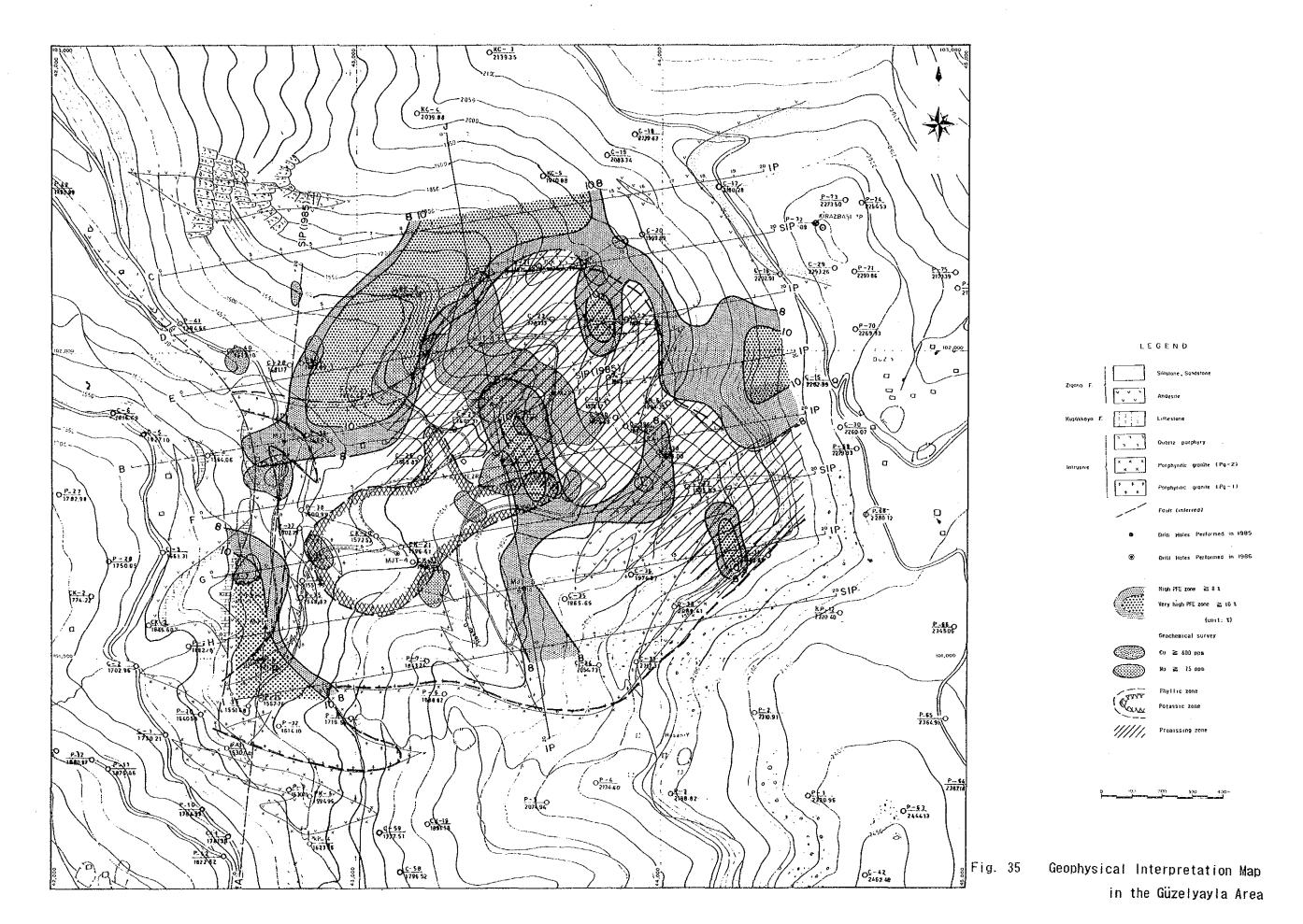
Table 4-2 Relationship between Cu, Mo% and Phase Spectrum

	SPECTRUM TYP					Έ	
Cu (%)	Α	В	C.	D	Е	X	Y
≤ 0.02 ≤ 0.04 ≤ 0.06 ≤ 0.08 ≤ 0.10 ≤ 0.12 ≤ 0.14 ≤ 0.16 ≤ 0.18 ≤ 0.20	3 2	1 1 1 2 1		1 6 1 1	1 1 1	1 1 1 1	2 2 1
≤ 0.22 ≤ 0.24 ≤ 0.26 ≤ 0.28 ≤ 0.30 ≤ 0.40	1	1		2			
Total No.	7	8		15	4	5	5

\ \ \ \ \ \ \ \ \ \ \ \ \ \ \ \ \ \ \	SI	PΕ	ТГ	RUN	<i>A</i> '.	ΓYΙ	PΕ
Mo (%)	Α	В	С	D	Е	Х	Y
≤ 0.002	5	4		6	2	2	1
≤ 0.004		1		2		2	1
≤ 0.006	1		:		1		2
≤ 0.008	1	1					
≤ 0.010				3		1	
≤ 0.012		1					
≤ 0.014				2			
$\leq 0.016$					1		
≤ 0.018				1			
≤ 0.020				1		1	
≤ 0.022							
≤ 0.024		1					
≤ 0.026							
. ≤ 0.028							
≤ 0.030							
≦ 0.040							
Total No.	7	8		15	4	5	5

Table 4-3 SIP Response and Cu. Mo% in the Classification of Phase Spectrum

) Type	Sample		Phase	PFE	Resistivity	Си	Mo
	No.		-mrad		ohm-m	b/ A	8/
Α	7	Min.	3.5	0.45	849	0.03	0.000
		Max.	63.0	9.30	10,068	0.21	0.008
		Ave.	24.3	3.70	4,336	0.09	0.002
В	8	Min.	17.5	2.60	797	0.01	0.000
		Max.	40.1	6.00	6,253	0.35	0.024
		Æve.	28.2	4.12	2,934	0.14	0.006
D	15	Min.	-4.3	-0.19	126	0.03	0.000
_		Hax.	114.0	20.20	7,300	0.40	0.020
		Ave.	32.8	5.70	1,688	0.13	0.007
E	4	Min.	8.5	0.40	195	0.02	0.000
		Max.	31.5	3.90	911	0.14	0.015
		Ave.	20,1	2.40	468	0.08	0.006
х	В	Min.	38.6	0.83	449	0.01	0.001
		Max.	541.2	144.57	1,795	0.12	0.019
		Ave.	307.3	70.80	1,066	0.07	0.005
Y	4	Min.	19.5	2.40	852	0.04	0.004
_		Max.	221.2	37.22	4,200	0.12	0.010
		Ave.	83.0	13.00	2,047	0.08	0.006



-75,76-

with specific metal contents of copper or molybdenum, but copper rich samples have a tendency to show a spectrum of A-, B- or D-type. Samples rich in molybdenum often show a spectrum of B-, D- or E-type. But, types of A, B, D and E are also common in samples of poor Cu and Mo content, making it difficult to presume a metal content by one of these types. X- or Y-type, which are popular in massive ores, has been observed in samples of low Cu or Mo content, but it is noticed that samples of these types have not been detected in samples with high Cu or Mo content. Samples of X- or Y-types in this area are megascopically rich in pyrite. Spectra of these types can be related with pyrite mineralization.

According to Table 4-3, phase spectra of high copper content are of B- and D-types and these are followed by A-, E- or Y-, and X-types in descending order. The copper content of B- and D-types are some 1.7 times larger than that of others. The phase stands at about 30 mrad with a value of 4 to 6% PFE. Molybdenum contents are highest with D-type, moderate with B-, E- and Y-type spectra and the lowest with A-type. Samples of X-type spectra are rather poor in Cu and Mo contents.

Consequently, samples with high copper content often show a B- or D-type spectrum with moderate values of phase, PFE and AR. Samples associated with X- or Y-type are related with pyrite mineralization with a low value of Cu or Mo.

Due to low values of metal content, copper and molybdenum did not give a specific spectrum. It is possible that the specific spectrum is obtained if samples have more metal content.

# 4) Conclusive remarks

As discussed above, the horseshoe shape zone of PFE higher than 10% is correlative with the pyrite shell in the phyllic zone and a target area for exploration is assumed to exist in this shell through to the zone of potassic alteration.

Interpretation Map (Fig. 35)

The survey result is summarized in the interpretation map. PFE anomalies areas with over 8 % and 10 % with respect to the plan map of PFE (N=1) and the promising area for copper and molybdenum mineralization, as is determined by simulation analysis and from the results of the laboratory test. The phyllic zone and potassic zone from the 1985 geologic survey and the geochemical anomalous zones of over 600 ppm for Cu and over 75 ppm for Mo are also marked on the same map.

#### Chapter 5 Drilling Survey

# 5-1 Outline of the Diamond Drilling

# (1) Purpose of the Diamond Drilling

As a result of geological and geochemical surveys carried out in the initial and second phases of the project, a porphyry copper type ore deposit was found as a promising target for future exploration in the Güzelyayla Area. In the third phase, a drilling survey consisting of five holes (total hole length 1,505m) was planned and subsequetly carried out in order to explore underground emplacement of the porphyry-copper type ore deposit, and to investigate and unravel the relationship between the emplacement conditions and the ore deposit and the results of geological, geochemical and geophysical surveys.

The Purpose of these cores are as follows;

MJT-4: Exploration in the center of the potassic zone

MJT-5: Exploration of the phyllic zone in the south-eastern Güzelyayla

MJT-6: Exploration of the molybdenum anomalous area found by soil

geochemical survey

MJT-7: Exploration of the IP anomaly area

MJT-8: Exploration of the SIP anomaly area

# (2) Outline of Drilling Operation

# (1) Location of drill holes

	Y	х	Z [m sea level]
MJT-4	43 131	01 338	1,578
MJT-5	43 550	01 227	1,857
мјт-6	43 482	01 640	1,635
MJT-7	43 639	01 860	1,752
MJT-8	43 409	02 023	1,761

### ② Drilling operation method

The wire line drilling method using an NQ type diamond bit as far as possible was applied. Drill inclination was vertical.

#### (3) Core survey

A geological columnar section 1/200 in scale was complied, and colour photographs of all drilling cores collected were taken.

### (4) Chemical assay of drilling cores

Whole cores collected were split along the core extention, and half

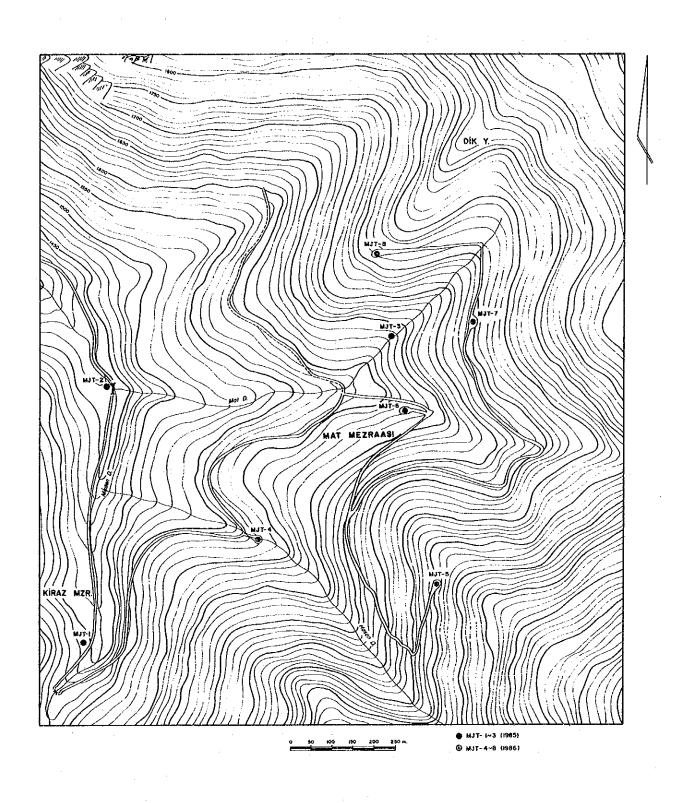


Fig. 36 Location Map of Drilling Holes in the Güzelyayla Area

pieces of the split core were chemically assayed to detect molybdenum and copper content for the enter section, while selected samples were analysed for gold, silver, tungsten, and tin content.

# (5) Laboratory works on the core

Microscope observations of rock thin sections and ore polished specimens, measurement of homogenization temperature and salinity of fluid inclusions, and detection of altered minerals by X-ray diffraction meter were performed.

# (3) Drilling Holes Performed

Drilling	Drill	Drill	Dip	Surface	Core	Core	Period
No.	length	length		soil	length	recovery	
	planned	performed	(°)			•	
MJT-4	300m	301.00m	-90	0.00m	300.70m	99%	June 24- July 10
MJ1-5	300m	301.00m	-90	0.00m	297.20m	99%	Sep.10 - Sep.23
MJT-6	300m	301.00m	-90	0.00m	301.00m	100%	June 24- July 12
MJI-7	300m	301.00m	-90	0.00m	300.80m	99%	July 25- Aug.10
MJT-8	300m	301.00m	-90	0.00m	298.90m	99%	July 25- Aug.12
Total	1,500m	1,505.00m		0.00m	1,498.60m	99%	

### 5-2 Drilling Operation

# (1) Drilling Method

The drilling operation was performed by means of the wire line method using a diamond drilling bit of NQ size (75m/m diameter). All drill sites had exposed bedrock at the surface.

Bentonite mud water was circulated during the drilling operation, and cutting oil (lubricant) was suitably added in the circulating mud water in order to reduce torque resistance caused by collapse in the hole.

Geology of the Güzelyayla area consists of andesite, porphyritic granite (Pgl and Pg2), basalt and quartz porphyry. At the predominantly alterated sections of rocks in the hole, the rocks are soft and brittle, and have many well-developed fractures (includes cracks, joints, and faults). This often

caused loss of circulating mud water and much flash water through fissures. Meanwhile, strong silicified rock was very hard to drill.

# (2) Drilling Machine, Equipment and Consumables

The Longyear L-38 was used for the drilling operation. Types and specifications of the machine, engine, pump and equipment, and amount of consumables are shown in Tables 5 and 7.

### (3) Operation Members and Shifts

The operation of move-in and move-out from site to site, and preparation work in the site were performed by a shift per day system, while the actual drilling operation was carried out by three shifts per day with eight working hours per shift. One drilling shift consisted of five members, a Japanese drilling engineer, a Turkish assistant driller (MTA) and three Turkish workers.

### (4) Transportation and Road Construction

The drilling machines, equipment and consumables were transported from the East Black Sea Regional Office of MTA located in Trabzon, to a place near these drilling sites by a large truck, and then to the drilling sites by a small truck. As there was no access road to individual drilling sites, a new 1.5 km road was constructed by bulldozer.

Drill	Length	Dozer	Compressor	Worker	Explosives	Light oil	Gasoline
(Period)	km	set	set		kg	e	e
MJT-4						<del></del>	
MJT-6	0.3+α	1	1 1	300	80	2,629	1,199
(20 May	~ 19 Jun	e)					
MJT-7							,
MJT-8	1.7	1	1	290	180	3,960	1,002
(20 June~ 18 July)							
MJT-5	0.8	1	2	570	510	6,416	2,516
(19 Jul	y∼ 6 Se	p.)					
Total	2.8+α	1	2	1,160	770	13,005	4,717
(20 May	~ 6 Sep	.)					

Dozer; Hanomag K-18 CE (Weight; 22 Ton) Compressor: Maksam 4.5 m<sup>3</sup>/m (Weight; 1,750 kg)

# (5) Water Supply

The water necessary for the drilling operation was run through a polyethylene pipe line from a neighbouring stream, but it was nessecary to pump up at the drilling sites of MJT-5 and MJT-8.  $^{\circ}$ 

### (6) Withdrawal

After completion of the third phase drilling survey, the drilling machines and equipment were stored in the storehouse of the MTA office in Trabzon.

#### 5-3 Results of the Diamond Drilling

# (1) MJT-4

The hole reached massive bed rock at 6.10m after cutting through the surface with an NQ size diamond bit, circulating dense bentonite mud water. After reaming by the NW casing shoe bit, NX casing pipes were inserted at 6.10m

Below 6.10m, NQ wire line method, bentonite mud water and cutting oil were used for the operation. The whole rock was altered porphyritic granite (Pg1).

The rock has undergone strong biotitization and sericitization from surface to 301m. Mineralization was weak between 85m and 150m. Below 150m, the rock gradually become more strongly silicified and mineralization became strong toward the deeper part. The drilling was completed at 301.00 m.

Depth (m)	0~ 301.00
Mud Water	Bentonite mud water
	Cutting oil
Bit Exchange (pcs)	NQWL bit(10)
Pump Pre. (kg/cm²)	5~15
Pump Feed ( $\ell/\min$ )	50
Pump deri (ℓ/min)	50
Bit Pre. (kg/cm²)	500~ 2,000
Bit Rot. (rpm)	300
Core Recovery (%)	99.9

# (2) MJT-5

The hole reached massive bed rock at 6.10m, after cutting through surface with an NQ size diamond bit, circulating dense bentonite mud water. NW casing pipe was inserted at 6.10m after reaming with the NX casing shoe bit.

Below 6.10m, the drilling operation was carried out by wire line method, circulating mud water mixed with bentonite and cutting oil. The rock between surface and 133.90m in depth was strongly sericitized pophyritic granite (Pg1). Below 133.90m, severely fractured chloritized basaltic andesite continued to 300m, and the last one meter (300~301m) intersected with sericitized Pg1 again. It was very difficult to drill because of the loss of mud water circulation along fractures at 37.5m, 64.5m, and 170m. The hole was completed at 301.00m.

Depth (m)	0~6.10	6.10~133.90	133.9000~ 301.00				
Mud Water		Bentonite mud water					
	Cutting oil						
Bit Exchange (pcs)	NQWL bit(1)	NQWL bit(4)	NQWL bit(7)				
Pump Pre. (kg/cm²)	5	5	10				
Pump Feed (ℓ/min)	50	50	50				
Pump deri (ℓ/min)	50	20~ 30	20~ 30				
Bit Pre. (kg/cm²)	2,000	2,000	2,000				
Bit Rot. (rpm)	300	300	300				
Core Recovery (%)	100	100	100				

# (3) MJT-6

As porphyritic granite (Pg1) was exposed at the surface of the site, the hole was drilled using an NQ diamond bit, circulating bentonite mud water, and was reamed with an NW casing shoe bit. NW casing pipes were inserted through porphyritic granite to 9.10m.

Below 9.10m, NQ wire line method, mixed mud water of bentonite and cutting oil were used for the drilling operation. The loss of water circulation was caused at 194m. The lithology of this drill hole consists of porphyritic granite of Pg1 and Pg2. The major rock was Pg1 with well-developed fissures and strong sericitization to 112.5m and from 244 to 289m, but Pg2 was weakly altered between 112.5 and 244m. Mineralization accompanied by copper and molybdenum mineralization occurred in Pg1. The drilling was completed at 301.00m.

Depth (m)	0~9.10	9.10~63.00	63.00~93.00	93.00~ 301.00
Mud Water		Bentonite mud	water	
		Cutting oil		
Bit Exchange (pcs)	NQWL bit(1)	NQWL bit(2)	NQWL bit(4)	BQWL bit(3)
Pump Pre. (kg/cm²)	. 5	5	10	15
Pump Feed (ℓ/min)	50	50	50	50
Pump deri (ℓ/min)	50	50	0	0
Bit Pre. (kg/cm²)	500~1,000	2,000	2,000	2,000
Bit Rot. (rpm)	300	300	300	300
Core Recovery (%)	100	100	100	100

### (4) MJT-7

The hole reached massive bed rock at 6.10m after cutting through surface with an NQ size diamond bit, circulating dense bentonite mud water. After reaming with the NW casing shoe bit, NX casing pipes were inserted at 6.10m

Below 6.10m, NQ wire line method, bentonite mud water and cutting oil were used for the operation. The whole rock was altered porphyritic granite (Pg1).

The rock has undergone strong sericitization from surface to 301m, and ribbon structure was predominant from 25 to 254m. Below 254m, it gradually become more strongly silicified and mineralization became poorest toward the deeper part. The drilling was completed at 301.00m.

Depth (m)	0~301.00
Mud Water	Bentonite mud water
Bit Exchange (pcs)	NQWL bit(10)
Pump Pre. (kg/cm²)	5∼ 15
Pump Feed (ℓ/min)	50
Pump deri (ℓ/min)	50
Bit Pre. (kg/cm²)	500~2,000
Bit Rot. (rpm)	300
Core Recovery (%)	99.9

# (5) MJT-8

As porphyritic granite (Pg1) was exposed at the surface of the site, the hole was drilled using an NQ diamond bit, circulating bentonite mud water, and

Table 5 Drilling Machine and Equipment Used (Güzelyayla Area)

Drilling Machine Model "L - 38"  Specifications:	2 set
Capacity	700 m ( BQ - WL )
Dimensions L $ imes$ W $ imes$ H	$2,150 \text{mm} \times 1,170 \text{mm} \times 1,450$
Hoisting capacity	4,500 kg
Spindle speed	Forward 236,490,900,1,510 rpm
Engine Model " F4L912 "	18 ps / 1,800 rpm
Drilling Pump Model " 535 RQ "	2 set
Specifications :	
Piston diameter	70 mm
Stroke	70 mm
Capacity	Discharge capacity 132 Q/min
	Max pressure 56 kg/cm²
Dimensions L × W × H	1,905mm $ imes$ $788$ mm $ imes$ $940$ mm
Engine Model "WISCON"	18ps / 2,000 rpm
Wire line hoist	Attached to drilling machine
Derick	Attached to drilling machine
Drilling tools	
Drilling rod	NQ - WL 3 m 234 pcs
Casing pipe	HX 1.5 m 4 pcs
	NX 1.5 m 1 pcs
	NX 3 m 21 pcs

Table 6 Drilling Meterage of Diamond Bit Used (Güzelyayla Area)

		m/pc													30.10							60.20	
-	MJT-8	E	30.15	34.30	33.30	30.70	28.50	31.90	22.90	30.20	35.80	23.25	`			64.45	64.00	60.40	53.10	59.05		-	6, 10
		Bit No	NN-20	NN-22	NN-24	NN-26	NN-28	NN-30	NN-32	NN-34	NN-37	NN-38				NR-22	NR-24	NR-26	NR-28	NR-30			A-12157
		ođ/ш													30.10							60.20	
	MJT-7	E	31.65	36.65	30.95	27.30	31.05	30.50	30.15	31,00	30.00	21.75		, , , , , , , , , , , , , , , , , , ,		68.30	58.25	61.55	61.90	51.00			6.10
		Bit No	-NN-19	NN-21	NN-23	NN-25	NN-27	NN-29	NN-31	NN-33	NN-35	NN-36				NR-21	NR-23	NR-25	NR-27	NR-29			A-12153
by Unit		эď/ш													33.44							60.20	
Drilling Meterage by Unit	9-TCM	E	30.50	32.10	36.40	38.35	26.75	26.75	36.35	35.60	38.20					62.60	74.75	53.50	71.95	38.20			00.6
Drilling		Bit No	NN-2	7 - NN	NN- 6	NN-8	NN-10	NN-12	NN-14	NN-16	NN-18					NR-12	NR-14	NR-16	NR-18	NR-20		1	A-12143
		m/pc			·	-									25.08					_		50.16	
	MJT-5	E	30.50	30.40	26.60	35.40	24.20	16.80	17.00	12.20	22.10	26.30	30.10	29.40		06.09	62.00	41.00	29.20	07.87	59.50		6.10
		Bit No	NN-39	NN-40	NN-41	NN-42	NN-43	77-NN	NN-45	NN-46	74-NN	NN-48	67-NN	NN-50		NR-31	NR-32	NR-33	NR-34	NR-35	NR-36		A-12163
		m/pc													33.44							60.20	
	4-TCM	E	32.00	30.50	24.40	36,60	33.55	39,20	30.95	42.70	31.10	· · · · · · · · · · · · · · · · · · ·				62.50	61.00	72.75	73,35	31.40			6,10
		Bit No	NN- 1	NN-3	NN- 5	NN- 7	6 - NN	NN-11	NN-13	NN-15	NN-17					NR-1 !	NR-13	NR-15	NR-17	NR-19			Casing shoe bit A-12138
	Size		!					Š											Ŋ				hoe bit
	Item							Bit	-						_				Reamer				Casing s

Table 7 List of Consumables Used (Güzelyayla Area)

						Quantit	у	
Discription	Specifi -cation	Unit						
:			млт-4	МЈТ-5	млт-6	MJT-7	MJT-8	<b>Total</b>
Light oil		Q	2,100	4,300	2,820	3,000	3,360	15,580
Petrol		Q	870	1,080	930	500	520	3,900
Engine oil		Q	40	100	40	50	80	310
Hydraulic oil		Q		40		40	20	100
Grease		Kg	20	30	20	20	20	110
Cement		Kg	30	30	30	30	30	150
Bentonite		Kg	2,000	11,200	6,000	2,950	9,425	31,575
C.M.C	***	Kg	55	90	80	40	180	445
Cutting oil		Q	40	140	100	80	240	600
Telstop		Kg	20	60	60	20	120	280
Diamond bit	NQ	pcs	9	12	9	10	10	50
Diamond reamer	NQ	pcs	5	6	5	5	5	26
Casing diamond shoe	NX	pcs	1	1	1	1	1	. 5
Core barrel Ass' y	NQ-WL	set	. 1	2	1	1	2	7
Inner tube	NQ-WL	pcs	2	4	3	2	3	14
Core lifter case	NQ-WL	pcs	10	12	12	8	10	52
Core lifter	NQ-WL	pcs	10	16	16	10	10	62
Thrust ball bearing	NQ-WL	pcs	10	10	10	8	8	46
Chack piece	NQ-WL	set	1	]	1	1	1	5
Cylinder liner	535-RQ	pcs	3	3	3	3	3	15
Valve seat	535-RQ	pcs	6	6	6	6	6	30
Steel ball	535-RQ	pcs	6	6	6	6	6	30
Piston rubber	535-RQ	pcs	6	6	6	6	6	30
Core box	NQ	pcs	- 59	59	59	60	63	300

Working Time Table of the Drilling Operation (Güzelyayla Area)

		Drilling		Shift	ı,	Working man	g man				Working Time	ime			
on of on			i <u></u>	· <del></del>		-			<u> </u>			4.14.	1 2 4 2 5 2	Road con-	
24-24-2	Bit	Drilling Core		Drilling	Total	Engineer	Worker	Drilling	Other	Recove-	Total	Removing	water traspor-	and	G.Total
	size	length	<u> </u>		\$				working ring	ring			tation	others	
		E	E	shift	shift	man	man	#	ત	,c	£	r.	Ę	'n	
4-TCM	ŊĠ	301.00	300.70	67	82	78	300	236.00	156.00		392.00	50.00		32.00	474.00
MJT-5	ŊŎ	301.00	297.20	77	80	126	588	205.00	131.00		336.00	48.00		32.00	616.00
MJT-6	ÒN	301.00	301.00	55	99	778	306	248.00	192.00		440.00	50.00		32.00	522.00
MJT-7	. Q	301.00	300.80	67	83	93	372	244.00	148.00		392.00	78.00		64.00	504.00
MJT-8	ŎΝ	301.00	298.90	55	67	e 6	372	244.00	196.00		00.044	78.00		78.00	536.00
Total	QN	1,505	1,498.6	250	332	7/7	1,938	1,177.00	823.00		2,000.00 244.00	244.00		408.00	2,652.00

Table 9 Record of the Drilling Operation of MJT-4

	Dr	illing len	gth	Tot	al	Sh:	ift	Workin	g man
	-		:		Core				
	Shift.1	Shift.2	Shift.3	Drilling	length	Drilling	Total	Engineer	Worker
	m	m	m	m	FD.	shift	shift	man	man
19 June	Pds						1	3	8
20 June	Pds	   					2	3	8
21 June	Pds						3	3	8
22 June	Pds						4	3	12
23 June	Pds <sup>.</sup>						5	3	12
24 June	1.50	;		1.50	1.20	1	6	3	12
25 June	3.05 ·	6.10	Ins-C.P	10.65	10.35	3	9	3	12
26 June	6.10	3.55	5.60	25.90	25.60	3 .	12	3	12
27 June	6.10	6.10	6.10	44.20	43.90	3	15	3	12
28 June	4.30	6.10	6.10	60.70	60.40	3	18	3	12
29 June	6.80	6.60	6.70	80.80	80.50	3	21	3	12
30 June	6.10	8.30	6.95	102.15	101.85	3	24	3	12
1 July	6.10	8.25	7.00	122.50	123.20	3	27	3	12
2 July	6.10	6.80	8.45	144.85	144.55	3	30	3	12
3 July	6.10	9.15	6.10	166.20	165.90	3	33	3	12
4 July	6.10	6.10	6.80	185.20	184.90	3	36	3	12
5 July	8.45	8.05	7.20	208.90	208.60	3	39	3	12
6 July	6.85	7.00	6.40	229.15	228.85	3	42	3	12
7 July	8.20	7.15	7.10	251.60	251.30	3	45	3	12
8 July	6.10	6,10	6.10	269.90	269.60	3	48	3	12
9 July	5.10	6.10	4.05	285.15	284.85	3	51	3	12
10 July	4.10	5.60	6.15	301.00	300.70	3	54	3	12
11 July	Dismant						<b>5</b> 5	3	12
12 July							56	3	12
13 July	!						57	3	12
·14·July	Dismant						58	3	12
	<u> </u>		-	-				· .	
Total	97.15	107.05	96.80	301.00	300.70	49	58	78	300

Road-con; Road-construction

; Preparation for drilling site

Transpor ; Transportation

Tra-Ress; Transportation and Reassemblage

Dismant; Dismantlement

Recoveri; Recovering work

Ins-C.P; Inserting casing pipe

Table 10 Record of the Drilling Operation of MJT-5

=	n <sub>n</sub>	illing len	ath	Tota	a1	Sh	ift	Workin	o man
	DI.	iiiing ten	gen	100	Core	511.		HOLKIII	5 mar.
	Shift.1	Shift.2	Shift.3	Drilling	length	Drilling	Total	Engineer	Worker
	m	m	m	m	m	shift	shift		man
17 Aug.	NH								
18 Aug.	NH								
19 Aug.	NH								
20 Aug.	Raod-con						1	3	12
-5	,						Ś	<b>\$</b> -	. \$
30 Aug.	Road-con						11	3	12
31 Aug.	Road-con	Road-con					13	3	24
\$							5	ş	S
6 Sep.	Road-con	Road-con					25	3	24
7 Sep.	Pds	Pds					27.	3	24
8 Sep.	Pds	Pds					29	3	. 24
9 Sep.	Pds	Pds					31	3	24
10 Sep.	7.30	6.20	7.40	20.90	20.90	3	34	3	12
11 Sep.	8.00	7.00	7.60	43.50	43.50	3	37	3	12
12 Sep.	8.40	7.10	5.50	64.50	64.50	3	40	3	12
13 Sep.	9.00	7.70	6.30	87.50	87.50	3	43	3	.12
14 Sep.	6.70	9.30	8.50	112.00	112.00	3	46	3	12
15 Sep.	8.10	7.50	7.70	135.30	135.30	3	49	3	12
16 Sep.	8.20	6.00	8.00	157.50	157.50	3	52	3	12
17 Sep.	6.90	5.60	8.10	178.10	175.00	3	55	3 <sub>.</sub>	12
18 Sep	6.90	8.10	7.10	200.20	196.90	3	58	3	12
19 Sep	7.70	7,30	7.30	222.50	219.20	3	61	3	12
20 Sep.	6.40	6.80	5.80	241.50	238.20	3	64	3	12
21 Sep.	5.80	6.20	6.50	260.00	256.20	3	67	3	12
22 Sep.	6.40	7.10	7.20	280.70	276.90	3	70	3	. 12
23 Sep.	7.70	6.50	6.10	301.00	297.20	3	73	3	12
24 Sep.	Dismant						74	3	12
25 Sep.	Dismant						75	3	12
26 Sep.	Dismant						76	3	12
27 Sep.	Dismant						77	3	12
Total	103.50	98.40	99.10	301.00	297.20	42	. 77	117	588

Road-con; Road-construction

Pds ; Preparation for drilling site

Transpor; Transportation

Dismant; Dismantlement

Recoveri; Recovering work

Tra-Ress; Transportation and Reassemblage

NH ; National holiday

Table 11 Record of the Drilling Operation of MJT-6

	Dr	illing len	gth	Tot	al	Shi	ift	Workin	g man
					Core				
	Shift.1	Shift.2	Shift.3	Drilling	length	Drilling	Total	Engineer	Worker
	m	n	m	m	m.	shift	shift		man
19 June	Pds						1	3	6
20 June	Pds					 	2	3	6
21 June	Pds				!		3	3	. 6
22 June	Pds						- 4	3	6
23 June	Pds						5	3	6
24 June	1.00			1.00	1.00	1	6	3	12
25 June	3.30	4.50	Ins-C.P	8.80	8.80	3	9	3	12
26 June	3.90	4.90	6.40	24.00	24.00	3	12	3	12
27 June	6.50	4.80	6.10	41.40	41.40	3	15	3	12
28 June	5.95	6.10	6.10	59.55	59.55	3	18	3	12
29 June	7.25	6.30	7.50	80.60	80.60	3	21	3	12
30 June	6.30	6.65	5.45	99.00	99.00	3	24	3	12
1 July	6.30	5,35	6.35	117.00	117.00	3	27	3	12
2 July	6.60	6.45	4.75	134.80	134.80	3	30	3	12
3 July	6.90	5.80	5.75	153.25	153.25	3	33	3	12
4 July	4.65	4.45	6.60	168.95	168.95	3	36	3	12
5 July	6.30	6.55	6.90	188.70	188.70	3	39	3	12
6 July	5.05	5.65	6.50	205.90	205.90	3	42	3	12
7 July	5.05	6.20	5.10	222.25	222.25	3	45	3	12
8 July	4.95	6.20	5.00	238.40	238.40	3	. 48	3	12
9 July	5.90	6.20	6.20	256.70	256.70	3	51	3	12
10 July	6.10	4.80	4.60	272.20	272.20	3	54	3	12
11 July	3.70	4.30	5.10	285.30	285.30	3	57	3	- 12
12 July	5.25	6.00	4.45	301.00	301.00	3	60	3	12
!	   Dismant						61	3	12
14 July	i ·		 				62 .	3	12
15 July	Dismant						63	3	12
16 July	ļ						64	3	12
Total	100.95	101.20	98.85	301.00	301.00	55	64	84	306
Abbassi	ation	L		· · · · · · · · · · · · · · · · · · ·	<u> </u>				· · · · · · · · · · · · · · · · · · ·

Road-con; Road-construction

; Preparation for drilling site

Transpor; Transportation

Tra-Ress ; Transportation and Reassemblage

Dismant; Dismantlement

Recoveri; Recovering work

Ins-C.P ; Inserting casing pipe

Table 12 Record of the Drilling Operation of MJT-7

		• • • • • •	. 1	m. A.	. 1	Ch	: 64	Workin	~ ====
	Dr.	illing len	gru	100	al Core	Sn:	ift	MOLKIII	R Marr
	Shift.1	Shift.2	Shift.3	Drilling	length	Drilling	Total	Engineer	Worker
	m	m	m	m	m	shift	shift	nan	man
	10								
15 July	Road-con						1	3	12
16 July	Road-con						2	3	12
17 July	Road-con						3	3	. 12
18 July	Road-con	ļ		i i	;		4	3	12
19 July	Pds						5	3	12
20 July	Transpor						6	3	12
21 July	Transpor						7	3	12
22 July	Tra-Ress				÷		8	3	12
23 July	Tra-Ress						9	3	12
24 July	Tra-Ress						- 10	3	12
25 July	1.60			1.60	1.60	1	12	3	12
26 July	4.70	2.60	5.55	14.45	14.45	3	15	3	12
27 July	5.15	6.05	6.00	31.65	31.65	3	18	3	12
28 July	9.00	6.00	8.70	55.35	55.35	3	21	3	12
29 July	6.00	6.95	7.50	75.80	75,80	3	24	3	12
30 July	9.10	8.35	6.00	99.25	99.25	3	27	3	-12
31 July	6.00	8.10	8.45	121.80	121.80	3	30	3	12
1 Aug	4.75	9.00	7.05	142.60	142.60	3	33	3	12
2 Aug	7.95	7.05	6.00	163.60	163.60	3	36	3	12
3 Aug	6.00	7.15	7.45	184.20	184.20	3	39	3	12
4 Aug	3.90	8.00	7.70	203.80	203.80	3	42	3	12
5 Aug	6.20	6.00	6.35	222.35	222.15	3	45	3	12
6 Aug	6.05	5.60	5.50	239.50	239.30	3	48	3	12
7. Aug	5.10	5.40	5.60	255.60	255.40	3	51	3	12 .
8 Aug	6.00	6.00	6.30	273.90	273.70	3 .	54	3	12
9 Aug	6.10	6.00	6.00	292.00	291.80	3	57	3	12
10 Aug	3.60	5,40		301.00	300.80	2	59	3	12
11 Aug	Dismant						60	3	12
12 Aug	Dismant					`	61	3	12
13 Aug	Dismant						62	3	12
14 Aug	Dismant						63	3	12
					<u> </u>				
Total	97.20	103.65	100.15	301.00	300.80	50	63	93	372

Road-con; Road-construction

Pds ; Preparation for drilling site

Transpor; Transportation

Tra-Ress ; Transportation and Reassemblage

Dismant; Dismantlement

Recoveri; Recovering work

Ins-C.P; Inserting casing pipe

Table 13 Record of the Drilling Operation of MJT-8

	T		<del></del>			,			
	. Dr	illing len	gth	Tot	al	Sh	ift	Workin	g man
		Ţ			Core		-		1
	Shift.1	Shift.2	Shift.3	Drilling	length	Drilling	Total	Engineer	Worker
	, m	m	m	m	m	shift	shift	man	man
		ļ							
17 July	Road-con			!			1	3	12
18 July	Road-con	İ	i i				2	3	12
19 July	Pds						3	3	12
20 July	Transpor				7100		4	3	12
21 July	Transpor				!		5	3	12
22 July	Transpor	[   			ı		6	3	12
23 July	Tra-Ress	1					7	3	12
24 July	Tra-Ress		·			, [	8	3	12
25 July	1.00			1.00	1.00	ן ו	9	3	12
26 July	5.45	2.95	4.35	13.75	13.75	3	12	3	12
27 July	6.00	5.50	4.90	30.15	30.15	3	15	3	12
28 July	5.10	7.15	7.35	49.75	49.75	3	18	3	12
29 July	4.60	6.40	3.70	64.45	64.45	3	21	3	12
30 July	6.30	6.00	9.00	85.75	85.75	3	24	- 3	12
31 July	7.50	4.50	7.80	105.55	105.55	3	27	3	12
1 Aug	6.00	4.20	5:50	121.25	121.25	3	30	3	12
2 Aug	5.70	6.20	6.80	139.95	139.95	3	33	3	12
3 Aug	6.30	6.30	4.40	156.95	156.95	3	36	3	12
4 Aug	5,60	5.80	6.20	174.55	174.55	3	39	3	12
5 Aug	4.80	6.60	5.90	191.85	191.85	3	42	3	12
6 Aug	4.90	4.90	5.40	207.05	206.05	3	45	3	12
7 Aug	4.70	5.50	4.50	221.75	219.65	3	48	3	12
8 Aug	5.30	5.20	4.30	236.55	234.45	3	51	3	12
9 Aug	5.40	6.00	6.60	254.55	252.45	3	54	3	12
10 Aug	5.70	5.50	5.50	271.25	269.15	3	57	3	12
11 Aug	4.90	6.30	5.60	288.05	285.95	3	60	3	12
12 Aug	6.30	4.20	2.45	301.00	298.90	3	63	3	12
13 Aug	Dismant	į					64	3	12
14 Aug	Dismant						65	3	12
15 Aug	Dismant						66	3	12
16 Aug	Dismant	i					67	3	12
1					İ				
Total	101.55	99.2	100.25	301.00	298.90	55	67	93	372
	iation	<b>/</b>	·			<del></del>			

Road-con; Road-construction

Pds ; Preparation for drilling site

Transpor; Transportation

Tra-Ress ; Transportation and Reassemblage

Dismant; Dismantlement

Recoveri; Recovering work

Ins-C.P ; Inserting casing pipe

Table 14 Summary of the Drilling Operation of MJT-4

	1	<del></del>	Survey	Period				Total	man day
	ļ. — -	Peri	od	Days	Work day	Off	day	Engineer	Worker
Operation					ďa	ys	days	ma	ın man
Preparation	on 19	June~2	3 June	5	5	Ì	-	15	48
					Drilling	- 1			
Drilling	24	June~1	0 July	17	17		-	51	204
	.				Recovering				
j	.							W. 4.	.
Removing	111	July~1	4 July	4	4		_	12	48
Total	19	June~1	4 July	26	26		_	78	300
rilling lengt	th				С	ore rec	overy	of 100 m l	nole
Length	3	00.00m		m					Core
planed			Overburden		Depth of ho	le	c	ore	recovery
Increase	3	01.00m		300.70m			rec	overy	cumulated
or		i			( m )		(	%)	(%)
Decrease	-	ļ	Core length					ļ	1.4
in								ļ	1. 1
length	-				0 ~ 10	0	-	99.8	99.8
Length			Core	%	100 ~ 20	0		100	99.9
drilled	3	01.00m	recovery	99.9	200 ~ 30	1	_	100	99.9
orking hours			h %	78			<del></del>		
Drilling		236.00	60	49.8	Effici	ency of	Drill	ing	
Other worl	king	156.00	40	32.9	Total m/wor	k		301.00m/1	7 days
Recovering	g				period(m	/day)		(17.7m/	day)
Total		392.00	100		Total m/total	al		301.00m/5	8 shifts
Reassembla	age	50.00		10.5	shift (m	/shift)		(5,18m)	shift)
Dismantler	nent	32.00		6.8	Drilling	length	/bit(e	ach sized	bit)
Water					Bit size	НХ		NQ	BQ
transporta	ation				Drilled				
Road const	truction				length			301.00m	
and others	S				Core			·	
G.Total		474.00		100	length			300.70m	İ
Casing pipe in	nserted		<u></u>						
		Meter	age						
Size	Meterag	e drill	ing × 100	Recovery					
		lengt	h <sub>.</sub>						
	( m )		(%)	(%)					
нх	: :								
N₩	6.1	2	.0	100				÷	
NW				- <del>-</del>	į				

Table 15 Summary of the Drilling Operation of MJT-5

P Drill L P I O D	ation Preparation Prilling Removing Fotal ling lengt Length planed Increase or	10 24 17 h	Peri Aug. ~  Sep. ~2  Sep. ~2  Sep. ~2  O0.00m	9 Sep. 3 Sep. 7 Sep.	24 14	Work day days 21 Drilling 14 Recovering	Off day days	Enginee m 63	er Worker nan mar 372
P Drill L P I O D	Preparation  Drilling  Removing  Fotal  ling length  planed  Increase  or	10 24 17 h	Sep.~2 Sep.~2 Sep.~2	3 Sep. 7 Sep.	14	21 Drilling 14 Recovering		63	372
Prill L o D i L d	Drilling Removing Fotal ling lengt Length planed Increase	10 24 17 h	Sep.~2 Sep.~2 Sep.~2	3 Sep. 7 Sep.	14	Drilling 14 Recovering	3		
R T Drill L P I O D I L d	Removing Fotal ling lengt Length planed Increase	24 17 h	Sep.~2 Sep.~2	7 Sep.		14 Recovering	-	42	168
R T Drill L P I O D I L d	Removing Fotal ling lengt Length planed Increase	24 17 h	Sep.~2 Sep.~2	7 Sep.		Recovering		1	100
Drill L p I Drill L p L d	Total ling lengt Length planed Increase or	17 h	Sep.∼2	· · · · · · · · · · · · · · · · · · ·	4	ļ		;	
Drill L P I O I L D I L D I d	ling lengt Length planed Increase or	h 31	-	7 Sep.		4		12	48
P I O D i 1 L	Length planed Increase or	3	00.00m		42	39	3	117	588
P O D i 1 L	planed Increase or		00.00m			Cor	e recovery	of 100 m	hole
I o D i L d	Increase or	3			m				Core
o D i 1	or	3		Overburden		Depth of hole	·	Core	recovery
i 1 1 d		, ,	01.00m		297.20m	And a series of the series of	re	covery	cumulated
i 1 L	Doorosea					(n)		(%)	(%)
l L	Dect enge			Core length			į	Ì	•
L	in						į		
L	length					0 ~ 100		100.0	100.0
đ	Length			Core	Z.	100 ~ 200		96.7	98.4
Worki	drilled	3	01.00m	recovery	99.0	200 ~ 301		99.5	99.0
	ing hours		A 1.	h %	%		····		
D	Drilling	. [	205.00	61	33.2	Efficier	cy of Dril	ling	•
0	Other work	ing	131.00	39	21.3	Total m/work	i	301.00m/	14 days
R	Recovering	<del>-</del>				period(m/d	lay)	(21.5m	ı/day)
Т	Total		336.00	100	54.5	Total m/total	:	301.00m/	42 shifts
R	Reassembla	ge	48.00		7.8	shift (m/s	hift)	(7.16m	/shift)
D	Dismantlem	ent	32.00		5.2	Drilling 1	ength/bit(	each sized	l bit)
W	Water					Bit size	HX	NQ	BQ
ŧ	transporta	tion			!	Drilled			
R	Road const	ruction				length	.	301.00m	
a	and others		200.00		32.5	Core		<u> </u>	
G	G.Total		616.00		100	length		297.20m	
Casin	ng pipe in	serted		<del></del>					
			Meter	age					
-	Size	Meterage	e drill	ing × 100	Recovery				
İ			lengt	h .		•			
		(m)		(%)	(%)	FIRE			
ļ	нх								
  -	NW	6.1	2	.0	100				

Table 16 Summary of the Drilling Operation of MJT-6

				Survey	Period	1			Tota	l mai	n day
			Peri	od ·	Days	Work day	(	Off day	Engine	er	Worker
Оре	ration					da	ays	days		nan	man
.	Preparation	on 19	June~2	3 June	5	5	Ĺ	_	15		30
1						Drilling					
	Drilling	24	June∼1	2 July	19	19			57		228
						Recovering				: "	
	Removing	13	July~1	6 July	4	4	-+		12		48
	Total	<del>-</del>	June~1		28	28	-	<del>-</del>	84		306
Dri	lling leng						Core	recovery	of 100 m	ho1	e
	Length	<del></del>	00.00m		m						ore
!	planed	-		Overburden		Depth of he	ole	į .	Core		covery
i	Increase	3	01.00m		301.00m			1	covery		nulated
	or		. , , , , , , , , ,			· (n)		•	%)		(%)
ļ	Decrease			Core length		,			ĺ		
	in	ļ	İ	<b>,</b>		9.1		Ì	<u> </u>		
	length	İ				0 ~ 10	00	<u> </u>	100		100
	Length			Core	%	100 ~ 20	00	<del> </del>	100		100
	drilled	3	01.00m	recovery	100	200 ~ 30			100		100
Мог	king hours	· · · · · · · · · · · · · · · · · · ·		h %	%						
	Drilling		248.00	56.4	47.5	Effici	iency	of Dril	ling		
	Other worl	cing	192.00	43.6	36.8	Total m/wor	rk		301.00m	/19 (	days
	Recovering	3				period(	n/day)		(15.8)	n/da	y)
	Total		440.00	100	84.3	Total m/tot	tal		301.00m	/64	shifts
	Reassembla	age	50.00		9.6	shift (r	a/shii	Et)	(4.70	n/sh	ift)
	Dismantler	nent	32.00		6.1	Drilling	g leng	gth/bit(	each size	d bi	t) ·
	Water					Bit size	Ю	K	NQ		BQ
	transport	ation		İ		Drilled				T	
	Road cons	truction				length			301.00m	_	·
	and other	3				Core					
1	G.Total	. <u></u> -	522.00		100	length _			301.00m		
Cas	ing pipe i	nserted									
			Meter	age	•					•	•
	Size	Meterag	e drill	ing × 100	Recovery						:
			lengt	h							
ļ		( m )		(%)	(%)						
ļ	нх				100				•		
	NW	9.1	2	.9	100						

Table 17 Summary of the Drilling Operation of MJT-7

4			Survey	Period				Total	man day
		Peri	od	Days	Work day	Off	day	Enginee	r Worker
peration		1, 1111			days		days	m.	an man
Preparati	on 15	July~2	4 July	10	10	ļ .	<del>-</del>	30	120
					Drilling				
Drilling	25	July~1	0 August	17	17			51	204
į					Recovering				
Removing	11	August^	~14 August	4	4		<del></del>	12	48
Total	15	July~1	4 August	31	31		-	93	372
rilling leng	th				Cor	e rec	overy	of 100 m	hole
Length	3	00.00m		m					Core
planed	ļ		Overburden		Depth of hole	:	C	ore	recovery
Increase	3	01.00m		300.80m			rec	overy	cumulated
or				]	( m )	Ì	(	%)	(%)
Decrease			Core length						
in	į			Ì				Ì	
length	1				0 ~ 100	<u> </u> !		100	100
Length	<u>-</u>		Core	%	100 ~ 200			100	100
drilled	3	01.00m	recovery	99.9	200 ~ 301			98.8	99.9
orking hours		J	h %	%		<u> </u>			
Drilling		244.00	62.0	48.4	Efficien	cy of	Drill	ing	
Other wor	king	148.00	38.0	29.3	Total m/work		!	301.00m/	17 days
Recoverin					period(m/d	lay)		(17.7m	/day)
Total		392.00	100	77.8	Total m/total			301.00m/	63 shifts
Reassembl	age	48.00		9.6	shift (m/s	hift)		(4.80m	/shift)
Dismantle		32.00		6.3	Drilling I	ength	/bit(e	each sized	bit)
Water	<u> </u>				Bit size	НХ	7	NQ	BQ
transport	ation		-	<u> </u>	Drilled		_		1
Road cons		32.00		6.3			ļ	301.0Óm	
and other					Core		_		<u> </u>
G.Total		504.00	1	100	length		İ	300.80m	
asing pipe i	nserted	<u> </u>							I
0 1-1-		_ Meter	age						
Size	Meterag	;		Recovery					
	<b>0</b>	lengt	•	<u> </u>					
	( m )	_	(%)	(%)	1				
нх			,	100	-				
NW NW	6.1		.0	100					
	ļ				-				

Table 18 Summary of the Drilling Operation of MJT-8

		T		Survey	Period				Total	l ma	n day
			Peri	od	Days	Work day	Off d	ay	Engine	er	Worker
0ре	ration					day	s d	ays	I	nan	man
•	Preparation	17	$July \sim 2$	4 July	8	8			24		96
						Drilling		İ			
	Drilling	25	July∼l	2 August	19	19			57		228
						Recovering					
	Removing	13	August^	-16 August	4	4			12		48
	Total	17	July~1	6 August	31	31			93		372
Dri	lling length					Co	re recov	ery of	f 100 m	hol	e
	Length	3	00.00m		m					C	ore
	planed			Overburden		Depth of hol	e	Cos	re	re	covery
	Increase	3	01.00m		298.90m			recov	very	cu	mulated
	or					( m )		( %	)		(%)
	Decrease			Core length							
	in					·					· 
	length					0 ~ 100		1	100		100
	Length			Core	%	100 ~ 200		~	100		100
	drilled	3	01.00m	recovery	99.3	200 ~ 301			97.9		99.3
Wor	king hours			h %	%						
	Drilling		244.00	55.5	45.5	Efficie	ncy of D	rilliı	ng		
	Other working	ıg	196.00	44.5	36.6	Total m/work			301.00m	/19	days
	Recovering					period(m/	day)		(15.8	m/da	у)
	Total	_	440.00	100	82.1	Total m/tota	1		301.00m	/67	shifts
	Reassemblage	:	48.00		9.0	shift (m/	shift)		(4.50	m/sh	ift)
	Dismantlemer	ıt	32.00		6.0	Drilling	length/b	it(ea	ch size	d bi	t)
	Water		· · · · · · · · · · · · · · · · · · ·			Bit size	НХ		NQ		BQ
	transportati	on.				Drilled			٠.		
	Road constru	ction	16.00		2.9	length			301.00m		
	and others					Core					
	G.Total		536.00		100	length			298.90m		
Cas	ing pipe inse	rted									
			Meter	age							
	Size Me	terag	e drill	ing $\times$ 100	Recovery						
			1engt	h							
		( m )		(%)	(%)						
	нх				100						
		6.1	2	.0	100	1					
	-		1		····································	1					

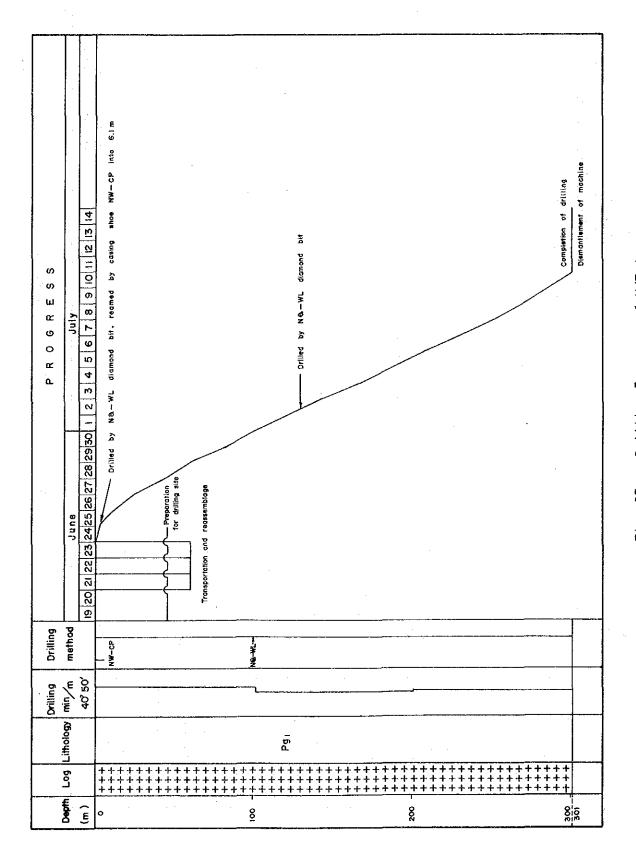


Fig. 37 Drilling Progress of MJT-4

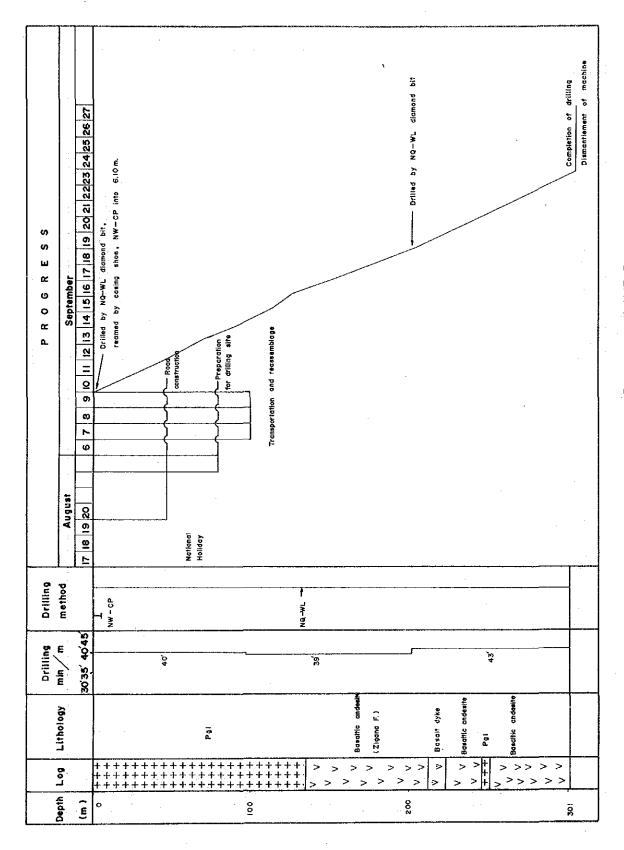


Fig. 38 Drilling Progress of MJT-5

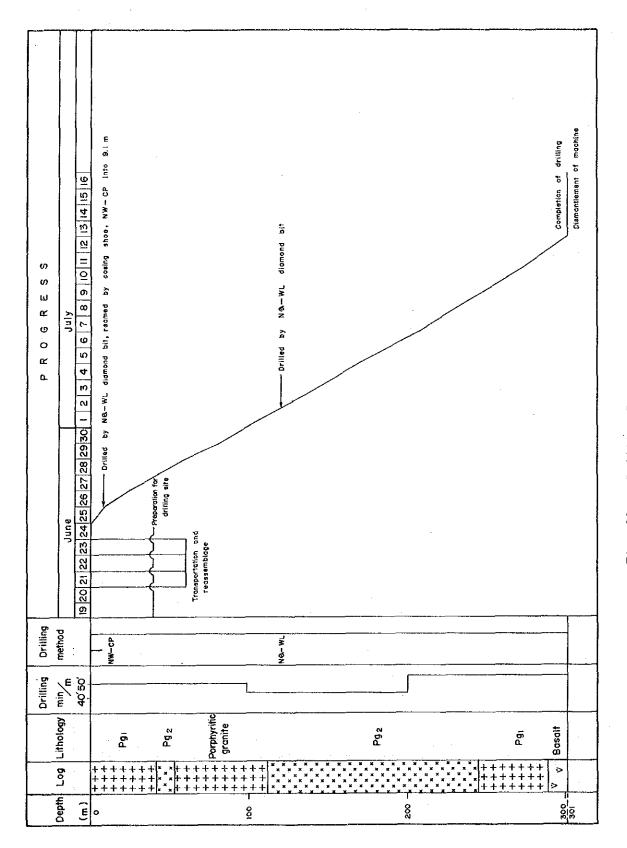


Fig. 39 Drilling Progress of MJT-6

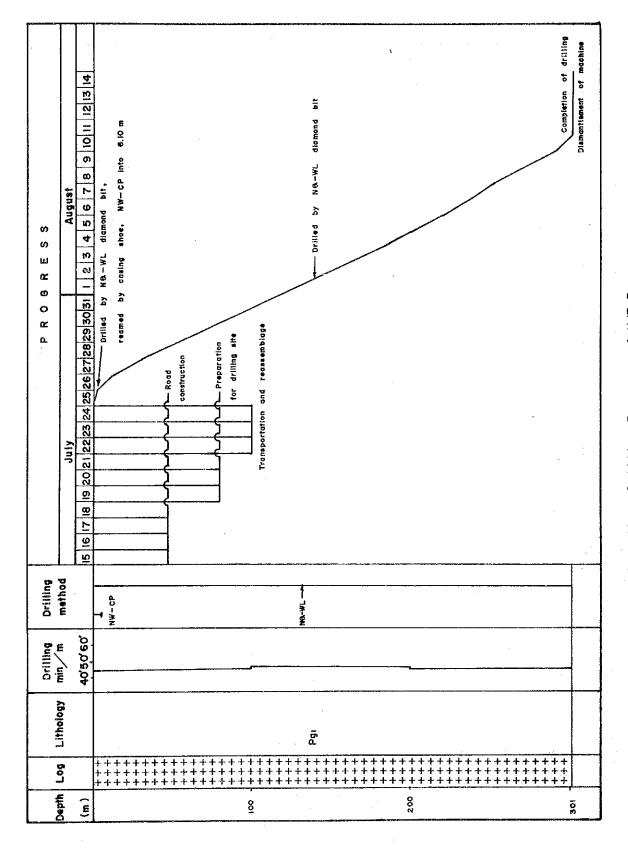


Fig. 40 Drilling Progress of MJT-7

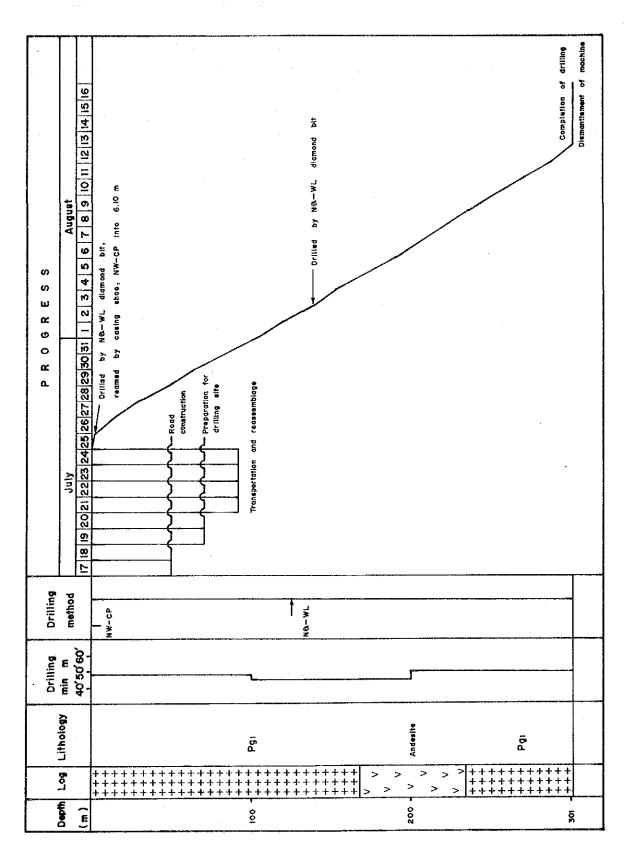


Fig. 41 Drilling Progress of MJT-8

was reamed by an NW casing shoe bit. NW casing pipes were inserted through porphyritic granite to 6.10m.

Below 6.10m, the drilling operation was carried out by NQ wire line method, circulating mud water mixed with bentonite and cutting oil. The rocks were strongly fractured from 84 to 234m, and collapse of the wall caused loss of water circulation. An especially sheared zone was from 112 to 123m in Pg1, and from 210 to 219m in andesite. The rock from surface to 169m in depth was strongly sericitized and chloritized and fractured porphyritic granite. Below 169m, severely fractured andesite continued to 234m, but below 234m the porphyritic granite underwent stronger biotitization and chloritization toward the deeper part. The hole was completed at 301.00m.

	*,			the state of the s
Depth (m)	0~93.00	93~114.00	114~ 224.00	224~ 301.00
Mud Water		Bentonite mud	l water	
		Cutting oil		
Bit Exchange (pcs)	NQWL(3)	NQWL bit(1)	NQWL bit (4)	BQWL bit(2)
Pump Pre. (kg/cm²)	5~ 10	5~ 10	5∼ 10	5~ 10
Pump Feed (ℓ/min)	50	50	50	50
Pump deri (ℓ/min)	50	20	0~10	10
Bit Pre. (kg/cm <sup>2</sup> )	500~2,000	2,000	2,000	2,000
Bit Rot, (rpm)	300	300	300	300
Core Recovery (%)	100	100	96	100

### 5-4 Geology and Mineralization of Drill Holes

#### (1) MJT-4

- 0.00~ 5.00m: Limonitized altered porphyritic granite(Pg1). It has undergone biotitization and sericitization.
- 5.00~ 30.00m: Dark brown Pg1(biotite-sericite). Colour changes gradually from dark brown to pale grey at depth. There are well-developed fissures in Pg1, and pyrite-chalcocite bearing quartz veins within 10mm width are embedded along with these fissures.
- 30.00~ 84.00m: Pale grey Pgl(biotite-sericite) accompanied by molybdenite and chalcopyrite along with fractures.
- 84.00~150.00m: Dark green Pg1(biotite-sericite-chlorite), and pyrites and chalcopyrites are embedded in fissures. Very minor amounts of disseminated chalcopyrite is partly observed

150.00~301.00m: Pale green biotitized and sericitized Pg1. Below 200m, sericitization and chloritization become stronger than biotitization. Molybdenites and chalcopyrites are embedded in fissures or accompanys quartz veins.

### [Alteration]

Porphyritic granite (Pg1) has undergone biotitization and sericitization throughout the hole from surface to hole bottom (301m). Small amounts of chlorite are also evenly contained. It is considered the center of the potassic zone as a result of the geological survey. The drilling survey indicated the same zone, but anhydrite was not observed in this hole as it was in the potassic zone of MJT-3.

### [Mineralization]

Mineralization accompanying pyrite, chalcopyrite and molybdenite is observed throughout from surface to hole bottom, but generally it is weak mineralization.

The range from 5m to 30m is regarded as the secondary enrichment zone because of the exsistance of chalcocite. Chalcopyrites and molybdenites are embedded in quartz veins and along fissures, while pyrites are disseminated in the rock and also in the quartz veins and along fissures. Comparatively strong mineralization occurs from 200m to 280.00m

# (2) MJT-5

- 0.00~ 10.00m: Limonitized and sericitized porphyritic granite (Pg1).
- 10.00~133.90m: White sericitized Pg1. Chalcopyrites and molybdenites occur not only in fissures but also as disseminations in the rock. Chalcocites are observed in the range from 10m to 112m, native copper at 109.6m.
- 133.90~211.70m: Dark green basaltic andesite. The basaltic andesite has been brecciated and undergone slight sericitization, but chlorite-epidote are the main altered minerals. The rock was so severely fractured that the cores were mostly broken and recovery was espectially poor from 170 m to 180m. Some chalcopyrite accompanies quartz veins and fissures. Magnetites are predominantly disseminated in the rock.
- $211.70 \sim 223.10m$ : A basalt dyke has intruded into the basaltic andesite.
- 223.10~246.00m: Dark green basaltic andesite is the same as that found in the range from 133.9 m to 211.7m. Some molybdenite accompanies quartz veins and fissures, very fine-grained chalcopyrites occur as disseminations.
- $246.00\sim250.80\text{m}$ : White and pale green aplitic Pgl. Sericites are mainly seen

as altered minerals. Molybdenites and chalcopyrites occur as disseminations in the rock.

250.80~300.00m: Dark green chloritized and epidotizated andesite. The ratio of distribution frequency per meter of quartz veins decreases slightly at depth, but emplacement of chalcopyrite and molybdenite along fissures is seen more predominantly.

 $300.00\sim301.00m$ : White and pale green aplitic Pg1 is the same as that in the range from 246m to 250.8m.

# [Alteration]

Porphyritic granite (PgI) has undergone sericitization, and small amounts of chlorite are also contained. Contrarily, basaltic andesite has undergone mostly chloritization and weak epidotization. Thus the alteration zone is classified into the phyllic and propylitic zones.

#### [Mineralization]

Mineralization with pyrite, chalcopyrite and molybdenite is emplaced in quartz veins and fissures. It also occures as disseminations in the rocks, but generally is weak. The range from 10m to 112m is regarded as the secondary enrichment zone because of the existence of chalcocite and native copper.

# (3) MJT-6

- 0.00~ 6.00m: Limonite-stained porphyritic granite (Pg1).
- 6.00~ 43.00m: White and pale green sericitized Pg1. Chalcopyrites and molybdenites occur not only in fissures and quartz veins but also as disseminations in the rock. Chalcocites are observed in the range from 6m to 23m.
- 43.00~ 53.00m: White and pale green porphyritic granite(Pg2). The rock has intruded into Pg1, and weak alteration of sericite and chlorite occurs. The boundary between Pg1 and Pg2 was not recognizable because of crushed cores.
- 53.00~112.50m: Sericitized Pgl with well-developed fissures. Chalcopyrites and molybdenites occur not only in fissures but also as disseminations in the rock.
- 112.50~244.00m: White and pale green Pg2. The rock suffers weak sericitization and chloritization. Very minor amounts of disseminated pyrite are observed. At some parts, there are small amounts of chalcopyrite and molybdenite along fissures. The boundary with Pg1 changes gradually.
- $244.00\sim289.00m$ : White and pale green Pg1. The rock has predominantly undergone sricitization, but chlorite increases towards the deeper

section. Potassium feldspar occures in the range from 244m to 250m, and anhydrite and gypsum occur in the range from 248m to 250m. Chalcopyrites and molybdenites were embedded along fissures and were also disseminated in the rock.

289.00  $\sim$  298.40m: Black basalt. The rock is massive and intruded at a low angle (aproximately 50°).

298.40 $\sim$  299.90m: Pale green Pg1 is the some as that in the range from 244m to 298.40m.

299.90~ 300.40m: Black basalt.

300.40~ 301.00m: Pale green Pg1. There is no mineralization.

### [Alteration]

The rock has undergone mainly sericitization from surface to hole bottom except for the intrusive rocks of Pg2 and basalt. Small amounts of chlorite are also evenly contained. Potassic feldspar begins to appear from around 244m in depth, anhydrite from around 248m in depth. Judging by this evidence, the alteration of the hole is divided into two zones, namely the phyllic zone from surface to 150m and the potassic zone from 150m to depth.

#### [Mineralization]

That is an oxidized and limonitized zone from surface to 1m. Thus sulphide minerals are not recognizable. The range from 1m to 23m, is regarded as the secondary enrichment zone because of the existance of chalcocite, but there are minor amounts of chalcopyrite and molybdenite disseminated in Pg1. Mineralization of chalcopyrite and molybdenite continues to 299.9m, but it tends to weaken toward the deep part. The mineralization is mostly emplaced as disseminations, along fissures and quartz veins.

# (4) MJT-7

### [Geology]

0.00~ 8.90m: Limonite-stained porphyritic granite (Pg1, sericite-chlorite)

8.90~254.00m: White and pale green sericitized chloritized Pg1. There are well-developed fissures; pyrite occurs along these fissures, chalcopyrites and molybdenites occur not only in fissures and quartz veins but also as disseminations in the rock. However, mineralization is very weak. Chalcocites are observed in the range from 5m to 42m. The rock texture is clearly ribbon structure in the range from 25m to 254m. Therefore, cores collected were plate shaped. Disseminated chalcopyrites and molybdenites are usually observed on the plates of the core, but they were of comparatively low grade.

254.00~301.00m: Pale grey Pgl. The rock suffered strong silicification and was collected as massive cores by the drilling operation.

Alteration minerals consist of biotite, sericite, gypsum and anhydrite. A minor amount of chalcopyrites and molybdenites can be found as disseminations or embedded in quart veins.

# [Alteration]

Porphyritic granite (Pg1) has undergone silicification and sericitization throughout the hole from surface to hole bottom (301m). Small amounts of chlorite are also evenly contained. Biotite-anhydrite-gypsum begins to appear from around 254m in depth. Judging by this evidence, the alteration of the hole is divided into two zones, namely the phyllic zone from surface to 254m and the potassic zone below 254 m.

# [Mineralization]

Mineralizations of pyrite, chalcopyrite and molybdenite are observed from surface to hole bottom, but generally are weak mineralization. The range of 5m to 42m is regarded as the secondary enrichment zone because of the exsistance of chalcocite. Below 42m, pyrites are disseminated in the rock and also in quartz veins and fissures, but minor amounts of chalcopyrite and molybdenite are embedded in quartz veins and along fissures. Comparatively strong mineralization occurs from 160m to 240m.

# (5) MJT-8

- 0.00~ 9.60m: Limonite-stained porphyritic granite(Pg1)
- 9.60~112.00m: White and pale green Pgl. Sericites are mainly seen as altered minerals, chlorites occur below 40m. Chalcopyrites and molybdenites occur not only in fissures and quartz veins but also as dissemination in the rock. Mineralization is weak. Chalcocites and covellines are observed along fissures in the range from 9m to 84m.
- 112.00 $\sim$  123.00m: Dark green chloritized Pgl. Although the rock is sheared, there is molybdenite in quartz veins.
- 123.00~169.00m: Dark green chloritized Pgl. The rock has mostly undergone chloritization and weak sericitization, chalcopyrites and molybdenites occur in quartz veins.
- 169.00~233.40m: Dark green andesite and pale grey Pgl. The rock has well-developed fissures. Chlorites are mainly seen as altered minerals. Chalcopyrites and molybdenites occur in quartz veins and along fissures. This range consists of andesite, intrusion of Pgl in the ranges of 195m to 196m, 198m to 199.75m

and 205,2m to 206m. Pg! has mainly undergone sericitization.

233.40~301.00m: Dark and pale green Pgl. Chlorites and biotites are mainly observed as altered minerals in the range from 233.4m to 283m.

Below 283m, sericites are predominant. Chalcopyrites and molybdenites occur in quartz veins and along fissures.

### [Alteration]

The rocks consist of pale green and dark green Pgl and andesite throughout the hole from surface to hole bottom (301 m). The alteration of the hole is divided into five zones as follows;

0.00∼ 40.75m	Pale green Pgl	Phyllic zone
40.75∼ 169.00m	Dark green Pgl	Propylitic zone
169.00~ 233.40m	Dark green andesite	Propylitic zone
233.40~ 283.00m	Dark brown Pgl	Potassic zone
283.00~ 301.00m	Pale green Pgl	Phyllic zone

# [Mineralization]

Mineralizations of pyrite, chalcopyrite and molybdenite are observed from surface to hole bottom, but generally is weak mineralization. The range from 9m to 84m is regarded as the secondary enrichment zone because of the existance of chalcocite and covelline. Below 84m, pyrites are disseminated in the rock and also in quartz veins and fissures, but minor amount of chalcopyrites and molybdenites are embedded in quartz veins and along fissures. Comperatively strong mineralization occurs from 230m to 300m.

### (6) Assay Result of Core

Drilling survey of the third phase resulted in five holes, totalling 1,505m in length. Whole core (1,498.60m) was split, and half of the split core was subjected to chemical analysis for two elements (Cu and Mo).

Taking into consideration that the ore deposit is porphyry copper type, the cores assayed were grouped and pulverized in each three meters. These samples (506 samples) were analyzed for Cu and Mo. Six samples which contained many sulphide ore minerals such as pyrite, chalcocite and covelline, had undergone strong biotitization and were regarded as scheelite by mineral-light. These six were selected to be analyzed for Au, Ag, Sn and W.

Average grades of these drill holes are as follows;

	Geol		Minera	lizati	on	Ī		·	Alt	erat	ion		
Depth	. 1	Lithology	Average grade				Alteration Zoning						
m	Log	MI CHOLOD)	Cu %		Cu+Mo	Bi	o S			Ep	An	Gy	
	+		0 - 20 m					<u> </u>	1				
-		Enrichment Zone	0.198	0.002	0.218			4	.j		i	i	
	+									-		٠	
-30	+												
	+		20-84m					 					
- 	* .						)		,				
- 60	+ +		0.072	0.002	0.092								
,	+												
-	+				!			• • •	-}	· • • • • • • • • • • • • • • • • • • •	••••••	······································	
	+ +		84-150m		Ī	]	İ		į.	ā.		•	
-90	+		0.4 (30)		Ī				-		-	•••••	
_	+								.i			*******	
	+ +								1				
-120	+	•	0.025		0.025			·	· <del>[</del>				
	+ +								}			•	
-	+								· [				
	+	•		l									
-150	+ +	Primary Zone	150 - 165 m								-		
_	+	PgI	0.069	0.003	0.099				ļ			•	
	+		165-301m						į				
-180	* +								. <del> </del>				
	+								į				
-	+	•			, 			}	Ī		*******		
-210	+								ļ			<u>:::</u> :	
	+ '						'		1				
-	+								. <b></b> .			<b>,</b>	
	+	•	0.082	0.005	0.132	'	•		1				
240	+ +	•				<sub> </sub> 			<b>†</b>				
	+								1				
	+ +				:								
-270	+		1.		: :				ļ				
	+								į				
-	+								ļ	(			
301	+								į				

Fig. 42 Geological Log of MJT-4

MJT-5

n t	Geol	Tithalana	Minera						tera Zonin			:
Depth	Log	Lithology	Averag Cu %			Rio	Sai			An	Gy	
m	+ +		0-9 m	110 %	GUITIO	DIO	1	1	L DP	- 1311		
	+	Leached Zone Pgl	0.015		0.015	ļj				I		
	+	·										
-30	+ .								• • • • • • • • • • • • • • • • • • •		•••••	
	+				ļ	ļ	1	į				
-	+											
	÷ '		:					1				
-60	+							}- 				
	+						1	1				
-	+	Enrichment Zone Pgl	9-105m		0,066							1
	+		0.000		0.000							
-90	+	·	•									
	+					<u> </u>		l.			· 	
	+	<b></b>						!				
-120	+	Primary Zone Pgl	105-135 m			\ \			*******			
120	+						-	į				
_	+	133,90	0.036	·	0.036			<u>i</u>	······			<b>,</b>
	v V		135~210 m					1	. !			
-150	v						•	}			<b></b>	
	V							- (	•			#
_	v										•••••	٠ <del>ڏ</del> ٠٠٠٠
	v	_ ,	0.087		0.087			- 1				tite.
-180	ν	Primary Zone Andesite							····		• • • • • • • • • • • • • • • • • • • •	Magne
	v .	(Zigana Formation)			j							<b>∑</b>
	V						****					
	v v	<u>211.70</u>			<u> </u>		• • • • • • •					<u> </u>
-210	v	Basalt dyke	210 - 222m									
	٧ 	223.10	0.014		0.014			1-	····-i		•••••	
]	v v	Primary Zone Andosite	222-246m						]			
- 240	٧ ٧	(Zigana Formation)	0.061	0.001	0.071			[.				
	+ +		246-301m		1		1	Ī	,			
-	۷ <sub>۷</sub>	250.80										
	v	Primary Zone Andesite						1	,			
- 270	v	Angesite (Zigana Formation)					·· • • • • • • • • • • • • • • • • • •				•••••	
	V		180.0	0.001	0.091	1						
}	, v	300.00						••••				
301	٧	Pg I					i	į			ļ	ł
	+ +		1		•				•			

Fig. 43 Geological Log of MJT-5

MJT-6

	Geol		Minera	lizati	on	Alteration
Depth		Lithology	Averag	e grad	e	Zoning
m	Log		Cu %	Mo %	Cu+Mo	Bio Ser Ch Ep An Gy
	+ +	Enrichment Zone Pgl	0-24m	0.032	0.724	
-30	+ + +	Primary Zone Pg! 43.00	24~45m 0.289	0.013	0.419	
) 	×	53.00 Pg1	45-53 in 0.046	0.001	0.056	
60	+		53-111 m			
	+ +					
-90	+ +	Primary Zone Pgi	0.294	0.011	0.404	
_	+	112.50				
-120	× × ×		li1-244m			
-	x x					
- 150	x x					·
-	×	Pgl	0.030		0.030	
-180	x x					
_	×					
-210	x x					; <b>\</b>
	х х х	·				
-240	+ +		244-289m			
-270	+ +	Primary Zone				
	+ +	Pgl	0.217	0.010	0.317	
301	v y	Basalt Dyke	289-301m 0,024	100.0	0.034	

Fig. 44 Geological Log of MJT-6

MJT-7

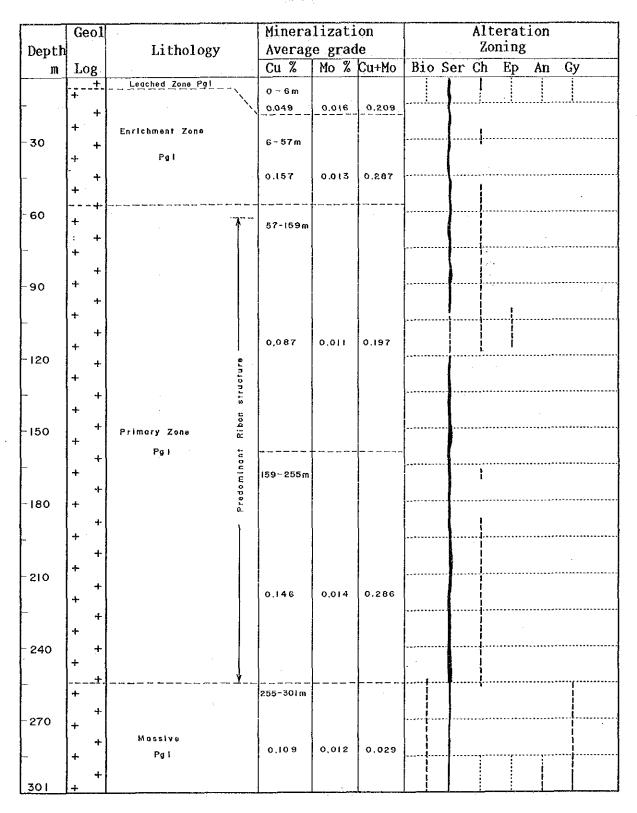


Fig. 45 Geological Log of MJT-7

8-TCM

	Geol		Minera	lizati	on		ببغسبين			tion		
Depth		Lithology	Averag	e grad	e	Zoning						·
m	Log			Mo %		Bio	Ser	Ch	Еp	An	Gy	
	+	Leached Zone	0-9m 0.044	0.022	0.262			]				
	+ .	\	9 – 5.4 m								:	
-30	+						}	. •			٠.	1
-	+ 	Enrichment Zone Pgl	0.264	0.007	0.334					-		,
-60	+ +	·	54-168m									••••
	+ +	:										•
-90	+ +					!					<u> </u>	
_	+ +					ļ <u>i</u>				· •	·	
-120	Δ	Primary Zone	0.162	0,012	0,282.							
_	+ +	Pgl				 						
-150	+	;			]		<u> </u>	 			:	•••
150	+							-			:	
<del>-</del> 	<del>:</del> -	169	168-225m									
-180	v v .										· · · · · · · · · · · · · · · · · · ·	•••-
_   	v V	Primary Zone Andesite	0.133	0.010	0.233	}	••••••	}-				
-210	v	(Zigana F)		-1+1*			****				·	
- ·	٧ - <del>-</del> ٧-	223.40	225-301m	<b></b>					••••			
-240	+						}					•
_    -	+ +											
- 270	+	Drimary Zooo	0.129	0.009	U 21a							
	+ +	Primary Zone Pgl	0.129	0.009	0.219		4		•			
301	+											

Fig. 46 Geological Log of MJT-8

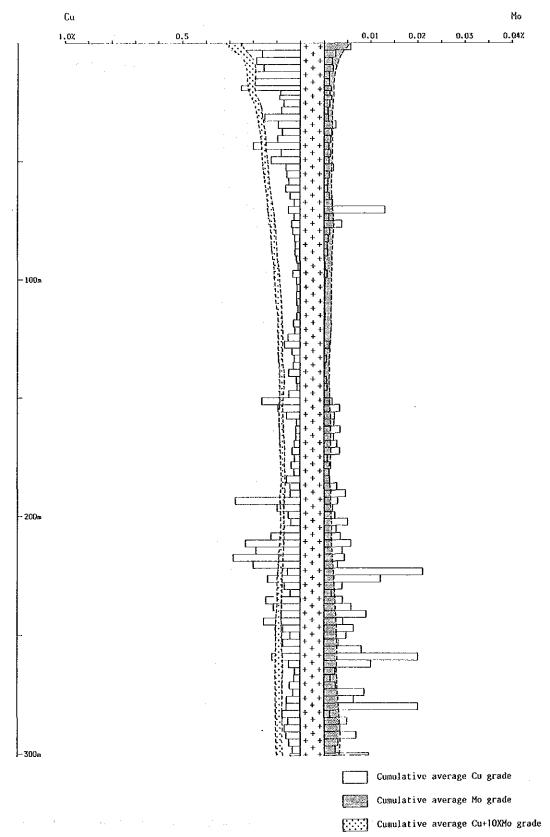


Fig. 47 Chemical Analysis Map of Copper and Molybdenum of MJT-4

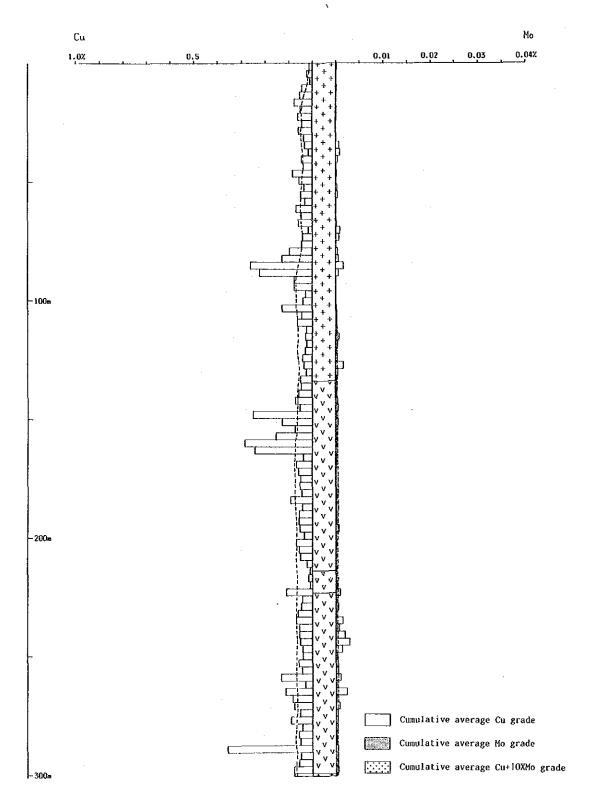


Fig. 48 Chemical Analysis Map of Copper and Molybdenum of MJT-5

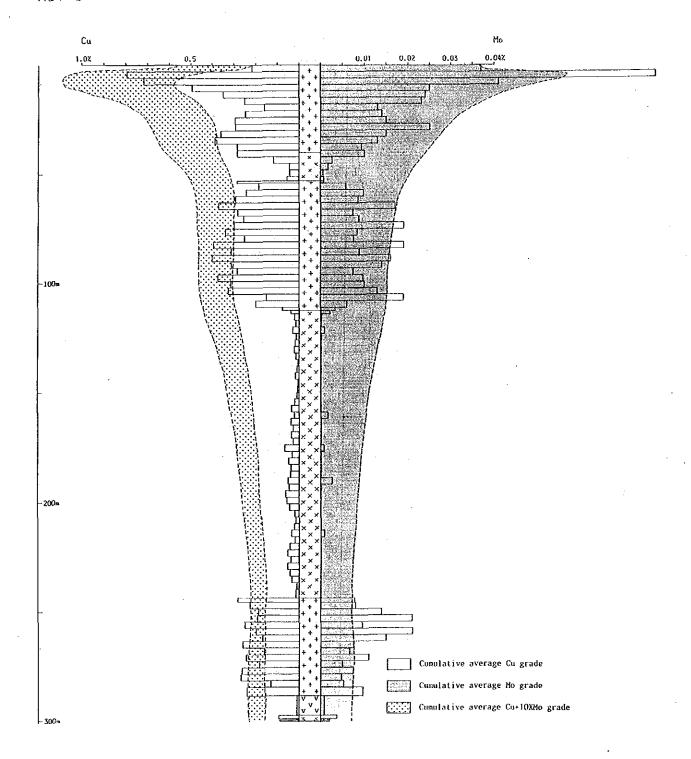


Fig. 49 Chemical Analysis Map of Copper and Molybdenum of MJT-6

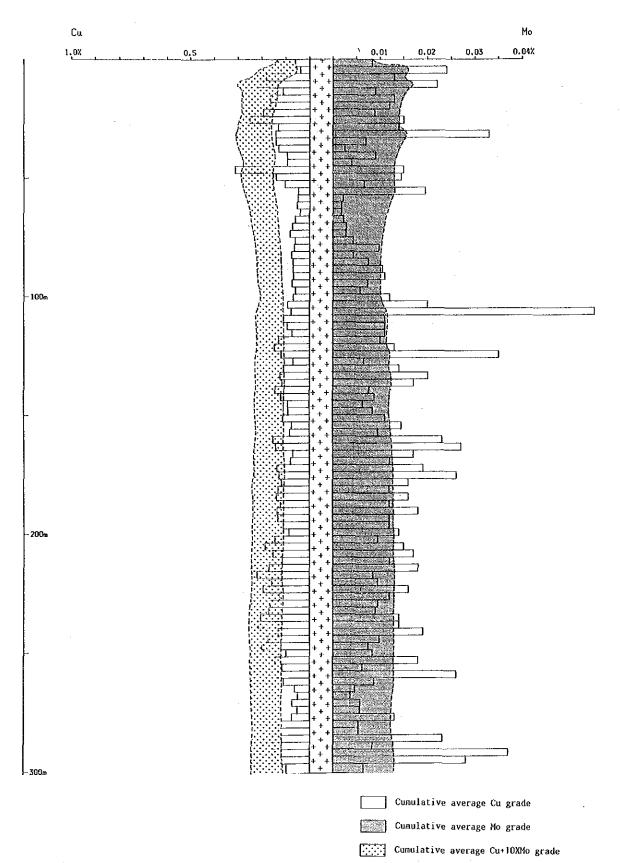


Fig. 50 Chemical Analysis Map of Copper and Molybdenum of MJT-7

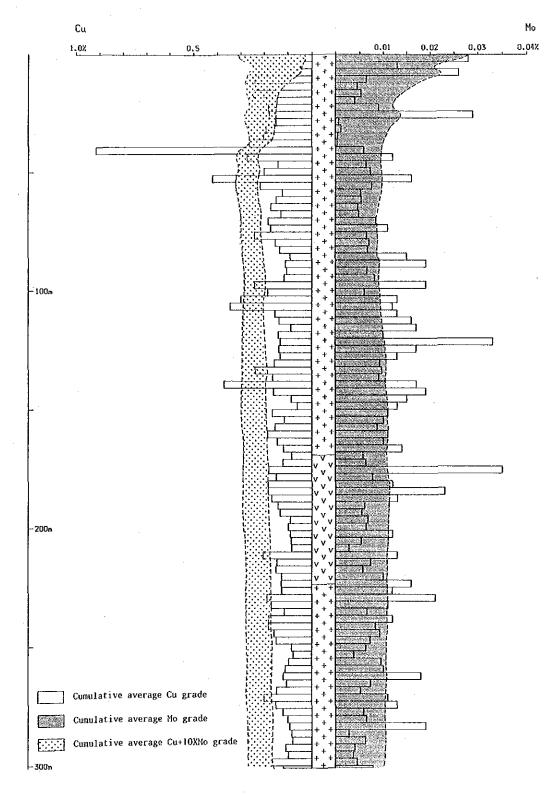


Fig. 51 Chemical Analysis Map of Copper and Molybdenum of MJT-8

Table 19 Cumulative Average Grade (No 1)

	MJT-4			i	MJT-5			t	AJT-6	
Sample No	Cu Mo	Cu+10×Mo	Sample	No Cu	Ho	Cu+10×Ko	Sample		Мо	Cu+10×Mo
	57.000	3050,000	740	124.000	2.000	144,000	437 438	2170,000 5080.000	370.000 570.000 •	5870.000 10760.000
336 2045	.000 40.500	2450.000	741	172.000	1,500	187.000	439	5766,670	516.667	10933.300
	).000 32.667 ).000 29.250	2306.670 2162.500	743	217.750	1.500	232.750	440	5560.000 5146.000	408.000	10060.000 9226.000
	25.600	2134.000	744 745	278.200 358.500	1,400	292,200 371,833	442	4710.000	378.333 342.857	\$493.330 7694.290
341 1978	.000 22.350	2151.670 2201.500	746 747	376.571 403.875	1.429 1.500	390.857 418.875	443	4265.710	317.500	7218.750
	.640 21.409 1,830 20.958	2087,730 2000,420	748	405.111	1.444	419.556	445 446	3922.220 3823.000	298.889 294.000	6911.110 6763.000
344 1668	.740 19.963	1868.370	749 750	421,100 415,091	1.400 1.364	435.100 428.727	447	3803.640	280,909	6612,730 8494,170
	.470 18.767 .970 17.879	1769.130 1753.760	751	406,583	1.833	424.917	448	3810.830 3736.920	268.333 254.846	6285,390
347 1520 348 1461	.640 18.472	1705.360	752 753	387.615 392.071	2.385 2.500	411.462 417.071	450	3673.570 3506.670	243,786 229,333	6111,430 5800,000
349 1425	.550 17.976	1644.260 1605.310	754 755	390.067 418.188	2,400	414.057	451 452	3315.000	216.125	5476,250
350 1462 351 1421		1636,960 1593,580	756	425,647	2.294	448.588	453 454	3145,180 3047.020	203,765	5182,820 5010,790
352 1410	.040 16.804	1578.080	757 758	420.889 423.842	2.333	444.222 448.053	455	3043.370	192.889	4972.260 4838.460
353 1365 354 1323		1535,410 1490,390	759 760	417.650	2.350	441.150	456 457	2980.560 2953.530	185.789 181,400	4767.530
355 1281 356 1250	.780 16.233	1444.120	761	430.143 431.045	2.227	453,318	458 459	2951.940 2986.850	176.857	4720.510 4752.300
357 1212	.440 15.576	1407.970	762 783	437.739 426.375	2.217	459.913 451.375	160	2980,460	172.043	4700.900
358 1170 359 1142		1327,150 1346,850	764	425.720	2.680	452.520	461 462	2963.360 2964.430	168.500 169.360	4648.360 4658.030
360 1107	.110 20.027	1307.370	765 766	426.654	2.615	452.808 473.630	463	2981.180	166.038	4641.560 4591.140
361 1078 362 1049		1285.510 1253.830	767 768	477.071 550.968	2.786 3.276	504.929 583.724	464 465	2964.470 2999.670	162.867 163.643	4636.100
363 1020 364 991		1220.150	769	606.933	3.367	003.043	466 487	3004,160 3037,360	161.034	4614.510 4647.360
365 965	.456 19.156	1187.180 1157.010	770 771	611,871 618,500	3.290 3.219	644.774 648.688	468	3040.020	160.323	4648.250
366 938. 367 912.		1124.620	772	606.000	3.152	837.515	469 470	3034.080 3055.470	157.625 155.788	4610.330 4613.350
368 894	.535 17.687	1071.400	773	599.206 618.657	3.088 3.057	830.088 649.229	471	3057.960	154,147	4599.430
369 872. 370 851.		1045,130	775 776	613.833 614.270	3.000 2.973	643.833 644.000	472 473	3063.730 3020.300	153.457 154.472	4598.300 4565.020
371 832 372 813	.435 16.324	995.676	777	604.421	2.921	633.632	474 475	2992,720 2963.150	151.919 150.333	4511.910 4466.480
373 794.	.860 15.518	972.847 950.035	778	595.974 586.825	3.026 3.050	626.231 617.325	476	2929.240	148.645	4415.690
374 778. 375 765.		929.803 913.933	780	579.585	3.000	609.585	477 478	2859.650 2792.030	144.885	4308.490 4204.900
376 752.	553 14.480	897.350	781 782	575.238 569.768	3.024	605.476 603.256	479	2731,150	138.061	4111.760
377 746. 378 745.	109 14.085	883 484 885.961	783 784	562,159 560,533	3.386 3.356	596.023 594.089	480 481	2669.270 2611.520	134.821 131.709	4017.480 3928.610
379 736. 380 725.		873.917	785	558.130	3.304	591.174	482 483	2557.190 2503.630	128.761	3844.800 3762.850
381 715.	580 13.341	860,963 848,986	786 787	558.170 561.542	3.277	590.936 594.458	484	2450.420	123.207	3682.480
382 710. 383 698.		841,560 828,215 .	788	560.694	3.327	593.959	485 486	2400.340 2353.730	120.628	3606.620 3535.090
384 687.	401 12.789	815.293	789 790	599.680 613.020	3.280	632.480 646.157	487	2306.530	115.745	3463.980
385 683. 386 701.		808.787 828.026	791 792	615.269 632.717	3.269 3.245	647.962 865.170	488 489	2261.740 2219.280	113.450 111.245	3396.240 3331.730
387 706. 388 793.	205 13.032	836.526	793	673.963	3.222	706.185	490 491	2180.450 2145.910	109.125 107.104	3271.700 3216.850
389 693.	475 13,346	835.742 826.932	794 795	705.891 699.804	3.182	737.709 731.411	492	2110.390	105.435	3164.740
390 684. 391 675.		821.352 813.435	796	899.193	3.140	730.597	493 494	2079.110	103.536	3114.480 3064,220
392 667.	626 14.064	808.269	797 798	697.052 693.373	3.068	728.086 724.051	195	2017.580	99.939	3016.970
393 561. 394 654.		805.868 787.345	799 800	690.483 686.541	3.033	720.817 716.705	496 497	1987.640	98.233 96.703	2969,970 2932,340
395 649. 396 643.		792.572 785.514	801	690.629	3.000	720.629	498 499	1938.050	95.108 93.566	2889.140 2849.940
397 642.	414 14.172	784.134	802 803	686.254 684.203	2.968	715.937 713.734	500	1889.540	92.089	2810,420
398 639. 399 635.		782.783 784.318	804 805	682.369 680.061	2.938	711.754 710.061	501 502	1865.780 1864,500	90.843 89.648	2772,210 2740,990
400 668. 401 673.	457 15.056	819.021	806	674.911	2.970	704.812	503		88.300 86.977	2706,200 2674,670
402 670.	965 15.189	824 111 822.856	807 898	675.059 673.391	2.956 2.928	704.618 702.867	504 505	1786.240	85.694	2643,180
403 667. 404 669.		824.064 827.773	809	670.914	2.900	699.914	506 507	1766.440 1743.700	84.449 83.239	2610.930 2576.090
405 677.	710 18.095	838.662	810 811	664.535 656.444	2.873 2.861	693.268 685.056	508	1721.070	82.064	2541,720
408 700.5		867.695 887.032	812 813	649.740 542.218	2.836	678.096 670.459	509 510	1699,990 1681.380	80.923 79.924	2480,610
. 408 746.5 409 763.5		919.813	814	648.587	2.907	677.653	511 512	1661.080 1644.330	78.842 77.791	2449,510 2422,230
410 760.1		938.194 960.951	815 816	645.513 643.169	2.882	874.329 872.390	513	1629,400	76.767	2397.070
411 768.:		982.386 983.394	817	842.808	2.897	671.782 673.848	514 515	1612.940 1598.870	75.770 74.798	2370.630 2346.860
413 783.	329 21.483	978.158	8 1 8 8 1 9	843.342 842.175	3.100	673.175	516 517	1583.760 1567.960	73.853 72.930	2322.280 2297.270
414 772.1 415 776.1		989.194 998.329	820 821	640.975 639.317	3.309 3.634	674.062 675.659	518	1549.740	72.031	2270.050
416 777.	144 22.996	1007.100	822	636.434	3.759	674.024	519 52 <b>0</b>	1526.660 1537.260	70.867 70.925	2235.330 2246.510
417 786.1 418 786.1		1018,600 1022,840	823 824	633.500 632.812	3.750 3.765	671.000 670.459	521	1546.090	71.034	2256.430
419 781.	698 23.960	1021.300	825	630 570	3.779	668.361	522 523	1549,940 1554,410	71.855 73.480	2268.490 2289.220
421 781.1	892 24.68?	1028.710	826 827	638.724 634.875	3,851 3,818	673.057	524	1565.640	73.731	2302.950
422 786.4 423 783.2		1053.670	828 829	640.663 642.989	4.045	681.112 683.211	525 526	1572.100	75.297 76.146	2333.560
424 777.0	086 27.479	1051.880	830	644.495	4.066	685.154	527 528	1583.770	76.133 76.020	2345.100
425 771.4 426 787.5		1044.830	831 832	643.141 646.085	4.065	683.793 686,710	529	1593.570	76.394	2357,510
427 762.5 428 760.5	953 27.920	1042.160	833	643.660	4.064	684.298	530 531	1596.360 1607.580	76.107 76.095	2357,430 2368,530
429 759.0	060 30.124	1060.300	834 835	642.884 641.365	4.032	683.200 681.469	532	1818.990	75.796	2376.950
430 759.6 431 756.7		1058.400	836 837	\$71.557 669.808	4.021	711.763	533 534	1615.630 1626.830	75.556 75.853	2371,200 2385,160
432 755.9	997 30.151	1057.510	838	868.849	4.020	709.051	535 536	1580,250 1576,950	73.834 73.654	2318,590
433 754,5 434 751.6		1059.930 1057.210	839 840	670.110 670.508	4.030	710.410 710.774	537	1574.480	73.545	2309.930
435 747.9	30.447	1052.380				,	538	1573.110	73.440	2307.510
436 746.8	370 30.661	1053.480								

Table 19 Cumulative Average Grade (No 2)

MJT-7

8-TLM

Sample	No	Cu	Мо	Cu+10×Mo	;	Sample	No Cu	ı Mo	Cu+10×Mo
639		.000	280.000	3090,000		539	600.0	000 83.0	00 1430.000
640		.000	205 000	2425.000		540			
641 642		.333 .750	223.333 183.500	2671.670 2763.750		541			
643	1203		155.800	2761.000		542 543			
644	1409		138.667	2795.830		544			
645	1413		124.571	2659,290 2665,630		545			
646 647	1463		120.250 139.111	2887.220		546 547			
648	1494		125.900	2753.500		548			
649	1531		115.545	2686.820		549			
650 651	1570		106.333 98.462	2633.750 2635.770		550			
652	2188		95.571	3143,930		551 552	1522.4		
653	2225		97.200	3197.000		553			
654 655	2173		95.063	3124.060 3100,290		554		130 131.00	
655 656	2183 2276		93.706 97.389	3250.280		555 556			
857	2270		96.211	3232.890		557	1513.1 1458.5		
658	2218		94.000	3158.250		558	1410.6	00 126.25	
659 660	2183		92.048 89.955	3103.570 3061.140		559 560	1369.1		
661	2122		88.130	3004.130		561	1324.8		
662	2110		87.958	2989.790		562	1268.9		
663 664	2094		88.840 87.923	2982.600 2985.190		563	1250.8	300 105.64	0 2307.200
665	2083		87.259	2956.480		564 565	1225.9		
688	2058	. 6 1 0	86.536	2921.960	4 - 1	566	1188.5		
667	2016		88.724	2903.450		567	1171.7	20 100.00	0 2171.720
668 669	1985 1955		92,100 91,226	2908.670		568 569	1156.0		
670	1930		90,906	2839.060		570	1110.8		
671	944	. 550	93,909	2883.640		571	1112.5		
672	1941		92,912	2870.880 2911.710		572	1100.4	40 98.97	1 2090.150
673 674	1972		93,971	2960.000		573 574	1095.1		
675	2000		95,649	2956.490		575	1085.9		
676	1982		97.342	2956.320		576	1082.3		
877 878	1954		99.205 99.225	2948.150 2932.000		577	1073.5		4 2213.330
679	1924		104.854	2972.680		578 579	1079.2		
680	1910		106,405	2974.760		580	1091.6		
681	1896		106.954	2966.050		581	1082.5	60 118,37	
682	1888 1899		106.636 106.422	2955.000 2963.780		582	1082.9		
684	1893		106.065	2954.570		583 584	1086.4		
685	1932		107.426	3007.020		585	1095.7		
888 887	1925		109,148 109,980	3017.290 3003.570		586	1097.9		3 2298.750
688	1877		110.380	2981.500		587 588	1093.9		
689	1873	. 240	110.373	2976.960		589	1090.6		
690 691	1859 1852		110.173	2960.870 2950.280		590	1084.0	40 118.57	
692	1852		109.741	2950.280		591	1079.11		
693	1845		109.564	2940.820		592 593	1087.50		
694	1833		110.107	2934,200		594	1086.5		
695 896	1815 1804		109,175	2888,530		595	1081.45		
697	1804		112,475	2929.240		596 597	1086.47		
698	1798		111.883	2917.750		598	1090.92		
699 700	1798		112.018	2919.750 2939.440		599	1094.51		7 2370.080
701	179B		114,175	2940.080	÷	600 103	1099.11 		
702	1792		113.328	2925.550		602	1105.08		
703 704	1785		112,415	2909.150 2888.640		603	1108,54		
705	1759	_	111,000	2869.550		604 605	1108.41		
706	1746		111.132	2858.090		606	1104.70 1109.93		
707	1733		110.290	2836.670		607	1120.80		
708 709	1720		109,114	2811.930 2819.510		608	1126.79		
710	1721		108.903	2810.630		609 610	1127.11		
711	1718		108,192	2800.340		611	1149.52		
712	1712		102 081	2792.910 2793.800		612	1155.34	0 128.595	2441.280
714	1699		108.921	2789.140		613 614	1165.80		
715	1702	.400	110,234	2804.740		615	1173.31		
716 717	1702 1695		110,231 109,658	2804.810 2792.090	•	616	1180.19	0 127.962	
718	1697		109,788	2794.940		617	1191.08		
719	1698	.700	109.457	2793.270		618 619	1203.06		
720	1697		109.244	2789.700		620	1211.04		
721	1694 1684		108.795 108.250	2782.710 2767.080		621	1220.54		2500.420
723	1673		107.424	2748.000		622 623	1217.92		
724	1665	.410	107.279	2738.200		624	1219.83		2501.350 2492.850
725 726	1658 1653		107.195 108.023	2730.860 2733.890		625	1221.55	0 128.828	2509.830
727	1646		107.629	2722.870	1	626	1220.17		
728	1644	. 500	107.011	2714.610		627 628	1213.60		
729	1648		107.044	2718.850		629	1200.71		
730 731	1646		107.293 105.763	2719.730 2709.620		630	1193.37	0 . 124.902	2442.390
732	1635	. 160		2698.240		631 632	1188.76		2438.330 2431.860
733	1627		107.189	2699.530		633	1190.26		
734 735	1619 1610		105.365 105.907	2682.970 2669.810		634	1194.84	0 124.552	2440.360
736	1605		105.235	2657.670		635 636	1198.09		2439.330
737	1598	. 910	104.545	2644.360		637	1201.99		2468.320 2485.300
738 739	1602 1601		103.950 103.864	2642.420 2640.150		638	1200.650		2473.940

Drill hole	Assayed range	Average grade						
		Cu% 、	Mo%	Equivalent Cu% *				
MJT-4	0.00~301.00m	0.075	0.003	0.105				
MJT-5	0.00~301.00m	0.067	· –	0.071				
MJT-6	0.00~301.00m	0.157	0.007	0.231				
MJT-7	0.00~ 301.00m	0.120	0.013	0.247				
MJT-8	0.00~301.00m	0.160	0.010	0.264				

<sup>\*</sup> Equivalent Cu% = Cu% + 10 X Mo%

Histograms of assayed results in each drill hole are presented in Figs.47 $\sim$ 51.

MJT-4 is situated in the center of the potassic zone. It has its highest grade in the interval 165m to 301m showing 0.005% Mo and 0.082% Cu. Other sections contain very low grades of Cu and Mo in spite of the existence of mineralization.

In the case of MJT-5, although 9.0m to 105.0m contains chalcocite and covelline as secondary copper, the grade is very low. The mean grade from 0.00m to 301.00m is 0.067% Cu, the grade of Mo is less than 0.001%.

MJT-6 was expected to have a higher ore grade because it was close to MJT-3, but Pg2 had intruded into Pg1. The mean ore grade is 0.487% copper equivalent in Pg1.

As mentioned already, MJT-7 and 8 are situated close to the center of mineralization, associated with phyllic to potassic alteration zones, and mainly characterized by well-developed fractures and dissemination type mineralization. Thus ore grade of MJT-7 and 8 were expected to be higher than those of MJT-4, 5, and 6, but they show 0.120% Cu, 0.013% Mo and 0.247% equivalent Cu in the range from 0m to 301m in MJT-7, and 0.160% Cu, 0.010% Mo and 0.264% equivalent Cu in the range from 0m to 301m in MJT-8.

The average ore grades of the secondary enrichment zone are as follows;

Drill hole	Assayed range	Average grade						
		Cu%	Mo%	Equivalent Cu% *				
MJT-4	0.00~ 21.00m	0.198	0.002	0.218				
MJT-5	9.00~ 105.00m	0.066	-	0.066				
MJT-6	0.00∼ 24.00m	0.404	0.032	0.724				
MJT-7	6.00~ 57.00m	0.157	0.013	0.287				
MJT-8	9.00∼ 54,00m	0.264	0.007	0.314				

<sup>\*</sup> Equivalent  $Cu\% = Cu\% + 10 \times Mo\%$