PART II DETAILED DISCUSSIONS

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PART II DETAILED DISCUSSIONS

Chapter 1 MJZC - 9

1-1 Progress of Drilling

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The location and the collar elevation of MJZC-9 are appended.

Summary of the drilling, record of the drilling operation and the drilling progress are shown in Tables 2-1-1 and 2-1-2, and Figure 2-1-1, respectively.

For the near surface zone to 40.00m, non-core drilling was 254mm percussion bit, and 219mm casing pipes were made by to 40.00m. At 40.00 to 72.00m, non-core drilling was inserted 203mm percussion bit, and 177mm casing pipes were made by inserted to 62.00m. Percussion drilling was continued by 165mm bit to 97m where percussion was given up due to the increase of flow out from 62m depth, changed to skid-mounted WL, water and 114mm casing pipes were inserted to 97m. Cuttings were collected at 1 m intervals during non-core drilling.

WL Coring was done by HQ bit to 201m, and NW casing pipes were inserted to 201m, drilled further to 1,144.76m by NQ bit and CHD rods. Rod grease, cutting oil and Drillprops were used in order to prevent vibration during operation.

Waters were supplied after pumping up from a water borehole drilled in the site.

Soft sheared zones were encountered at 405.8-420.2m. At the zone, the wall was cemented in order to prevent collapse.

Borehole deviation was measured at 60m intervals. Measurement showed northeastward deviation (Appendices).

1-2 Geology and Mineralization

The geologic log is appended. The geology of this borehole compared to that of the survey area described in 3-2 of PART I,

Table 2-1-1

Summary of the Drilling Operation on NJZC-9

79618 - 797 ann a 179 - 1897 a' 19			Survay Perl	od			1	Total Nan	Day	
		Period	1	Day	Work Day		Off Day	Engineer	Norker	
Operation										
Prepara	tion	06.07, 1995~	10. 07, 1995		-				· .	
		15.07, 1995		6		5		15		
Oritiin	e l	11. 07, 1995~	13.07,1995		Ortilling	53		224	. 43	
		17. 07, 1995~	21.09, 1995	71	Recovering	5	3	29		
Dismant	ling	14.07 1995								
		22 09 1995~	26.09, 1995	6		5	1	18		
Total				83		69	14	286	- 51	
Drilling	Length	'n	1	n	Core Recover	y of R	X0m Hola			
Length	Planed	1100.00	Overburden	60.00		1		Core		
Increas	e/Decréase	44.76	Core Length	1031.27	Depth of Hole		Core	Recovery		
in Lèr	eth .			·			Recovery	Cumulated		
Longth	Drilled	1144.76	Core		(m)		(1)	(\$)		
(N/C	Drilling)	97.00	Recovery	98.43	0.00~ 100.00		98. 20	98.20		
(Core	Drilling)	1047.76			100.00- 200.00		\$4.16	94.28		
forking ł	fours	h	5	×	200.00- 300.00		95.31	94.79		
Dr HHis	18	625.00	45.69	35.95	300.00- 400.00		97.99	95.85		
Other 1	lorking	\$50.00	40.20	31.65	400.00- 500.00		97.34	96.22		
Recover	ing	193.00	14.11	11.10	500.00- 600.00		100.00	98.97		
Subtota		1358.00	100.00	78.71			100.00	97. 47	· •	
Reassee	nblaga	32.00		1.84	700.00- 800.00		99.15	97.71		
Dismani	tlenent	18.00		1,04	800, 00- 900, 00		100.00	98.00		
Water S	Supply	270.00		15.54	900.00-1000.00		100.00	\$8.22		
Road Co	onstruction	8.00		0.46	1000.00-1100.00 99.60			98.36		
Transpo	ortátion	42.00		2 42	Efficiency (of Oril	ling			
Grand	lotal	1738.00		100.00	Total Length /		n .	day	n/day	
Casing Pi	ipa Inserted		· · ·		Drilling Period		1144.76	11	16 12	
		¥sterage	1		Total Length /		'n	shift	∎/shift	
Siza	Noterage	Deliling	Length	Recovery	Total Drilling		1144.75	100	11.45	
		- ×10	0	·	Shifts	;	:			
	(n)	<u>()</u>)	(1)	Drilling Ler	ngth / I	Each Bit (m)			
21960	40.00	3, 49		0.0		Drilled	Length	Core Le	ngth	
17766	62.00	5.42		0,0		· ·	40.00	-		
11450	97.00	8.47		0.0			22.00		<u> </u>	
h7	201.00	17.55		100.0		· · ·	35, 00		N,	
8 X	0.00	0.00		•	H0		104.00		98	
	·				NQ	·	943.76		933.	
					80		0.00		0.0	

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Table 2-1-2 Record of the Drilling Operation on MJZC-9

sta	Drilling Ler	gth (4)	Tatel	Dally fotal (core	Shife (shift)		Man Working (as	·?	Abbr
: 1	shift I	shift 2	Total Constated	Longth	Length	Or HILING	Total	Engineer	Norker	Band
15	Rd-con		0.00	0.00	0.00	0.00	1,00	1.00	4.00	· :36
7	PJs	-	0.00	0.00	0.00	0.00	1.00	3,00	10.00	Cen :Cen
8	Tra		0.00	0.00	0.09	0.00	1.00	3.00	2,60	<pre>ces</pre>
9	Dey off	Day of	0.00	0,00	0.00	0.00	0.00	3,00	11.00	i icas
0	Tra	20.00	50,00	53,00	0.00	2 00	2.00	6,00	10 00	Dist
12	30,00	17.00	\$7,00	37.00	0.00	2.00	2.00	1.03	10.00	:Dis
3		Drl-w/h	\$7.00	10.00	0.00	1.00	2.90	4.00	10.00	0r∎-c
1	Orl-w/h		97.00	0.00	0.00	0.00	1,00	3,00	6 00	. ≑Ore
5	Tre	•	97.00	0,00	D, 00	0.00	1.00	3.00	11 90	Orl e
5	Dey off	Dey off	97.00	0.00	0.00	0.00	0,00	2.00	2.00	Zdri Exc
17	9,25	18,60	120,85	27.65	27.35	2.00	2,00	3.00	8,00	Enc
18		Meint :	134,55	9.71 17,19	8.40 14.73	2.00	2.00	4.00	8 DG	las.
20	5.42	11.77	179.55	27.80	26,37	2.00	2.00	4.00	8.00	: Ins
ถ้า .	10.50		291,00	21,45	21,25	2.00	2.00	6.00	8.00	Ú BC
22	LA-CP	In-cp	201,00	0.00	0.00	2.00	1.60	4.00	8,00	: : : :
23	Day off	Day off	201.00	0.00	0.00	0.00	0.00	2.00	2.00 8.00	Masin. ∶:Mai
14	1.15		205_85	1,86	4,80	2.90	2.00	4,00	8,00	Out
75 ·	8.00	15.38	279.24	23 38	21.28	2.00	2.00	6,03	8.00	:16
26	14.92	9.93	254,09 266,81	24 85	24, 47	2.00	2,00	1.00	8.00	Pds
27	5.07	7.65 Dev. of (265.81	0.00	0.00	0.00	0.00	2.00	2.00	:Pr
28	Day off Day off	Day off Day off	255.81	0.00	0,00	0.00	0.00	2.00	2.00	ed
30	Dev off	Day off	255_81	0.00	0.00	0.00	0.00	2,00		1100
31	11,10	17,85	295,76	28 95	27.05	2.00	2.00	4.09		Rd-co
1	17.37	18.53	331.75	35.00	34,23	2.00	2.00	¢.00		:90:
2	17.00		358,76	25.00	24.80	2.00	2.00	4.00 4.00	6_00 6.00	Reas Reas
3	5,00		385,75	23.00 28.00	29,00	2,00	2,00	4.00	8.00	Reast
<u>.</u>	23, 12		413,76		7.70		2.00	4.00	8.00	:Re
5	Day off	Dev oft	421,76	0.00	0.00	0.00	0.00		2.00]	Reco
7		Cea	427,57	5.41	5.47	1,00	1.00	3,00	5.00	:30
1	Dr-ces	7,59	435,15	2.59	7.25		2.00	4.00		Sorv
5	16.60		459.75	34.60	34,31		2.00			: 5
10	11.00		195,62	25.85	25.20		2,00			Tea Tea
11		Racov-bt	435.05		0.4		5.00			Tash
12	Racov-ot	Racov-ot	495.05	0.00	0.0		0,00			: fa
13	Dey off	Day off Ces	495.05		9,00		2.00			¥ad g
14	Wesh Dev-off	Duy-olf	495.05		0,04		0.00			
15	Dey off	Dey off	495,05		0.00		0.00			Nt-b
1)	Osy off	Day off	495.05		0.00		0.00			: 1
18	Dey off	Dr-cen	435.05		0.0		1.00			
13	Ora-ces	Dr-cse	495.0				2.00			
20	Dr-ćen	Day off 3.76	496.0				2.0			
21 22	Dr-cee 23.9		544,90		45.1		2.00		D_00	
23	20.7					2.00	2.00			
21	6.0						2.0			
25	21.0						2.00			2
25	12.0						2.00			
27	18.0						7.0			
28	24.0						2.0			
29	12.0					2.00	2.0	¢_0	0 0.00	
31		O Day ell	799.7	6 6 .00	6,0	0 1.00	1.0			
ep 1	Day off	Day olf	799.7				0.0			
2	Day of I	Only off	857,8				0.0			
3	Dey off	Dey off	799.7				0.0			
4	Out-In-rd	12,00					2.0			
\$	21,4						2.0		0 B.00	
6	19.2					1 2.00	2.0	4.0	0 B.00	
	Qut-in-rd	Vash	919.7		0.0	2.00	2.0	6.0		
9	12.0		\$40,7	5 21.0	21.0		5.0			
10	18.0	0 12.00	\$70.7				2.0			Í
11	24.0		1012.7				3.0			
12	12.0			5 15.8	0 15.8 0 8.0		2.0			1.
13	2.3									1 ·
14	Qut-In-rd	Bash 4.76	1035.0							11
15	Wash Walnt	10.7					2.0			
17	11.1					3 2.00	2.0	0 6.0	8.00	}
18	8.					1 2.00	2.0	0.0	8.00	1
19	14.			6 25.3	2 25.5	2 2.00	2.0	0 4.0		
20	12.	17 14.5	1138.7	\$ 27.0						
21	1.	1.3								
22	Surv	Out-cp	1144.1							
23	Dis.	Day off	1111.1							
24	Day off	Day off	1144.7							
25	Dit	Day off	3144.7							
26										

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lacks the "Footwall Conglomerate" immediately below the "Ore Shale Horizon", but otherwise it agrees well. Geological description of the borehole is as follows.

Lower Roan Group

"Footwall Quartzite": 1,112.20 to 1,144.76m. It mainly consists of quartzite and sandstone with intercalation of biotite-anhydrite rock and thin conglomerate. the upper pebbly quartzite and coarse quartzite partly contain pelitic parts. Anhydrite, biotite and dolomite occur in the lower quartzite. The conglomerates are composed of quartz, altered granite and biotite schist pebbles and biotite matrix. Dissemination of minute chalcopyrite is observed at 1,112.20-1,121.40m.

"Ore Shale Horizon": 1,079.50 to 1,112.20m. It mainly consists of dolomitic argillites with grey thinly laminated lavers. The basal part of this formation is made up of schistose argillite. The interval of 1,080.20 to 1112.20m is the sulfide zone composed of chalcopyrite and mineralized pyrite, partly rich in chalcopyrite. The sulfides occur as dissemination in the bedding planes and dissemination in silica concretion or dolomite veinlet. Results of ore assay are shown in Table 2-6-The cobalt mineral of this borehole was identified to з. be Cattierite and carrollite (Table 2-5-2).

"Hangingwall Quartzite and Argillite": 1,069.60 to 1079.50m. It mainly consists of grey quartzites. In the upper part, pelitic bands are intercalated. The lowest part is composed of thin alternation of pelitic beds and sandy beds.

"Interbedded Argillite and Quartzite": 1,038.80 to 1,069.60m. It is mainly composed of dark grey pelitic to dolomitic sandstone with intercalation of thin argillite, dolomite and quartzite. The lower part is composed of quartzitic sandstone. Lenses of anhydrites are observed frequently.

"Upper Quartzite": 1,028.00 to 1,038.80m. It is composed of grayish white coarse quartzite with minor amounts of pelitic bands and thin dolomitic sandstone beds.

Upper Roan Group

"Interbedded Argillite, Dolomite and Quartzite": 989.20 to 1,028.00m. The upper part consists mainly of pelitic sandstone with intercalation of thin argillite and dolomite beds. The

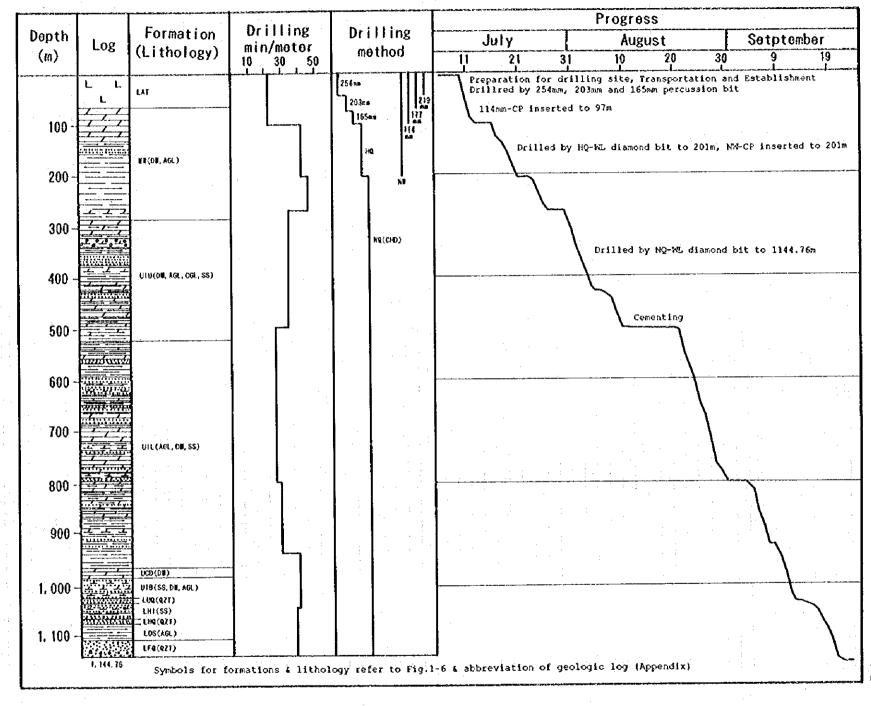
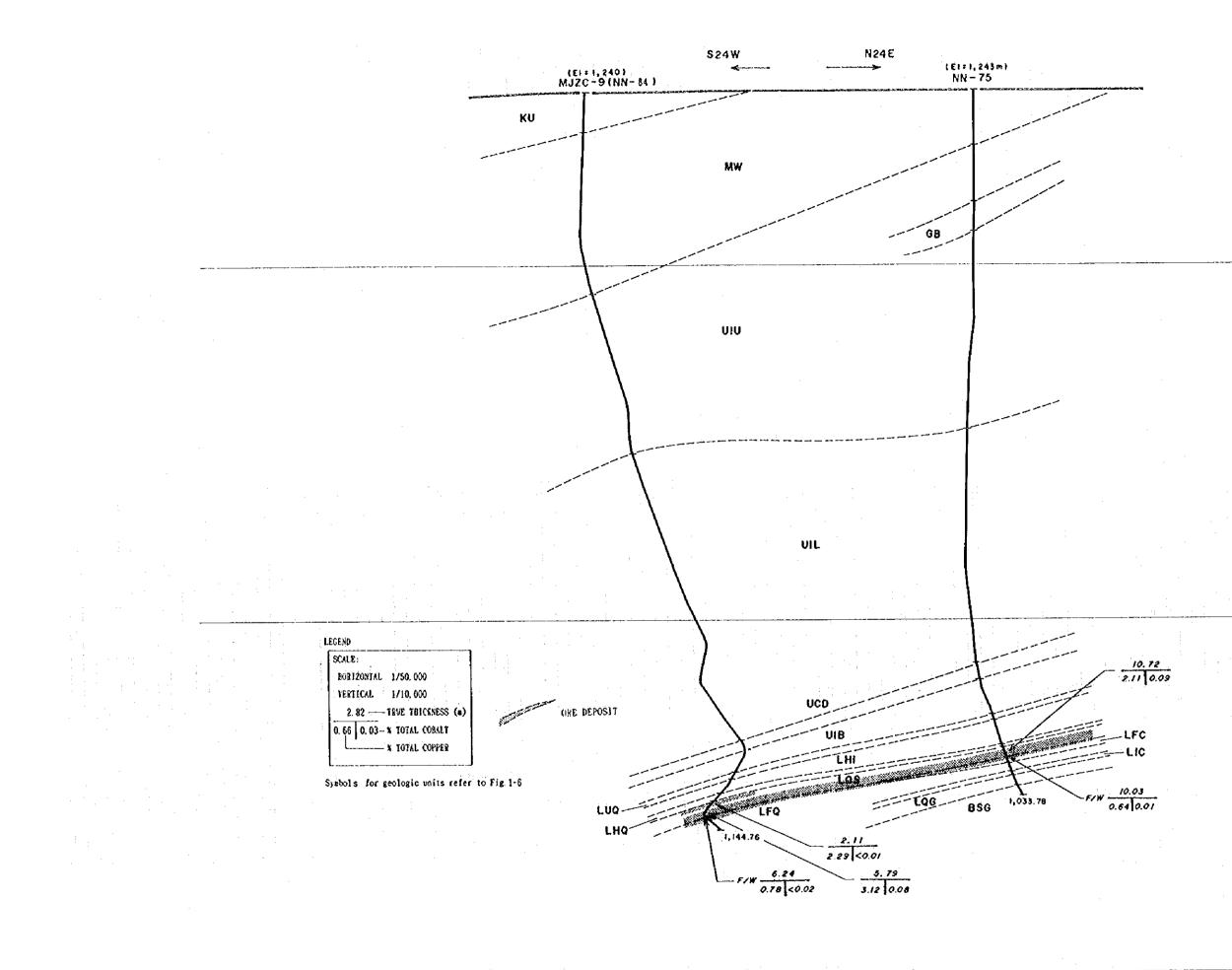


Fig. 2-1-1 Drilling Progress of MJZC-9

Taking out CP Dismantlement, Removing

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500 m. S.L.



Fig. 2-1-2 Geological Profile of Drill Hole (MJZC-9) - 61 - 62 - lower part consists of thin-bedded alternation of dolomite, argillite and sandstone with quartzite lenses.

"Cherty Dolomite": 969.20 to 989.20m. It mainly consists of massive white dolomite with silica lenses. In the upper part, grey argillite (Marker Shale) is intercalated. Very finegrained chalcopyrite and pyrite are disseminated between 984.60 and 987.70m.

"Arenite, Argillite and Dolomite with Anhydrite": 520.70 to 969.20m. It mainly consists of sandy argillite, dolomite and pelitic sandstone. The upper part, the middle part and the lower part are rich in pelitic sandstone, dolomite and sandy argillite, respectively. There are many lenses or patches or veinlets of anhydrite in this formation.

"Interbedded Argillite and Dolomite with Tectono-Breccias": The upper part consists of an alternation 283.30 to 520.70m. argillite and conglomerate, the middle part of of dolomite, alternation of sandy rock, dolomite and argillite, the lower part of alternation of dolomite and argillite. The conglomerate composed of dolomite pebbles of irregular shape, and 15 is Contemporaneous brecciations are developed in the argillized. argillite and sandstone, and many fractures are filled with dolomite or quartz. Minor amounts of gypsum bands and patches of anhydrite or disseminations of pyrite are partly observed in the upper part and the middle to lower part, respectively.

"Mwashia Group": 60.00 to 283.30m. The upper part is composed of pelitic dolomite. The lower part is composed mainly of argillite. The argillite is rich in shaly part with sandy or dolomitic intercalation.

1-3 Discussions

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NN-75 is located to the north of the present site in question. This NN-75 confirmed the occurrence of a high-grade copper deposit. The mineralization encountered in both holes is developed from the "Ore Shale" to the footwall and belongs to the chalcopyrite zone. The geology of MJZC-9 is shown in the cross section (Figs. 1-7, 2-1-2). The "Ore Shale" is thicker in this borehole than in NN-75. In NN-75, the ore shoot occurs as one unit in "Ore Shale", whereas in this hole, the shoot is

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separated into two within the "Ore Shale" and the lower one continues into the footwall quartzite. The highest grade of copper and cobalt in both holes occurs in the lower part of the shoot, and it is possible that they are continuous.

Basement depth contour of the area was prepared from three holes (MJZC-6, 7, 10) drilled in the vicinity. And it is inferred that both holes are located over a basement depression which extend in the NE-SW direction. MJZC-6, 7, 10 are located on or near palaeo-basement highs at the time of ore deposition (2-3, 2nd year report).

From the similarity of mineralization encountered at this hole and NN-75 and the geologic structure of the area, the ore shoot confirmed by the two boreholes is concluded to be a continuous ore deposit emplaced over a basement depression which extend in approximately NE-SW. And this depression was formed on a limb of the palaeo-basement high at the time of ore deposition.

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2-1 Progress of Drilling

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The location and the collar elevation of MJZC-10 are appended.

Summary of the drilling, record of the drilling operation and the drilling progress are shown in Tables 2-1-1 and 2-1-2, and Figure 2-1-1, respectively.

the near surface zone to 24.00m, non-core drilling was For 254mm percussion bit, and 219mm casing pipes were made by to 24.00m. At 24.00 to 32.00m, non-core drilling was inserted 203mm percussion bit, and 177mm casing pipes were made by inserted to 32.00m. Percussion drilling was continued by 139mm bit to 60m where percussion was given up due to the increase of flow out from around 32m depth. Then the method was water changed to truck-mounted WL, and 114mm casing pipes were inserted to 60m. Cuttings were collected at 1 m intervals during non-core drilling.

WL Coring was done by HQ bit to 173.90m, and NW casing pipes were inserted to 173.90m, drilled further to 663.90m by NQ bit and CHD rods. Rods were broken at 520.35m depth, became impossible to drill further, the hole was wedged (1.5°) at 454.92m and drilling continued.

Soft sheared zones were encountered at 490-512m and 598-At the zone, the wall was cemented in order to prevent 608m. and bentonite was used to drill further. At 958.40m collapse. depth drilled, rods breakage, hole bending and guide rod breakin casing pipes happened at above sheared zone, and the age drilling below 501m became impossible. These accidents seemed be caused by the collapse. Consequently the hole was again to (1.5°) at 448.13m and drilling continued to 628.76m by wedged NQ bit and BW casing pipes were inserted. Then core drilling was done by BQ bit to 1,009.86m.

Rod grease, cutting oil and Drillprops were used in order to prevent vibration during operation.

Waters were pumped up into a tanker from a water borehole

	Period			 A second sec second second sec				
			Day	Work Day		Off Day	Engineer	Worker
							1	
ion	04.07, 1995~1	07. 07, 1995				· · · ·	1	
	10.07, 1995							
	06. 10. 1995~	07 10 1995	6.00	ļ	5.00	1 00	21.00	62.0
1						1	· · · · · · · · · · · ·	821.0
				line ein T			and a second	398.0
	and the second		120 60		44.00	0.00	190.001	393.0
		27 11 1933	139.30				er e de la composición	
ing	•		· · · ·				i	
	• • • • • • •				1.1		-	
	28.11, 1995~0	01 12 1995		I I I I I I I I I I I I I I I I I I I		l i i i astriji	and the state of the second	49.0
			151.00		139.00	12.00	602.00	1330.0
	n		n	Core Recoi	very of 1	iDOm Hole		
lared	1100.00	Overburden	7.00				Core	
/Decrease	-90, 14	Core Length	924.68	Depth of Hole		Core	Recovery	
th						Recovery	Cumutated	
rilled	1009.86	Core		(m)		(1)	()	
rilling)	60.00	Recovery	97.35	0.00-100.00	3	96.45	95.45	
Drilling) 🗄	949,85		1997 - 1997 - 1997 - 1997 - 1997 - 1997 - 1997 - 1997 - 1997 - 1997 - 1997 - 1997 - 1997 - 1997 - 1997 - 1997 -	100.00- 200.00	 D	95.46	95.74	
	h	5						
	634,00	and a second				to and the second second second		
	and a second second		and a starty call.	·			an a san isis isa isa	
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	and the second second	and a second on a character of	• • • • • • •		and the second second	 I and a subset of the second se		a kanda sa
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	the second states	· · · · · · · · · · · · · · · · · · ·					97.35	
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and the second	3594.00	1 · · · · · · · · · · · · · · · · · · ·	100,00	Total Length /	1	a a	day	n/day
e Inserted		· · · · · · · · · · · · · · · · · · ·		Drilling Perio	bd	1009.86	151.00	6 69
1.1.1.	1			Total Length /	$t_{ij} = 0$	R I	shift	m/shift
Waterage -	 Orilling 	Length	Recovery	Total Brilling	đ i	1009.86	167.00	6.05
	×100			Shifts	4 - E			
(m)	(1)		(8)	Drilling (erigth /	Each Bit (m)		
24.00	2.38		0.00	Bit Size	Drille	1 Length	Core Lengti	
32.00	3.17			254mm	1			N/(
117.00	11.59		0.00	203mm		 A second second 		
173.90				and solar is only a set	• •			N/
111 <u>-</u> II			and the second second	and a second second second				110.0
VEV. 34	V2 - 2U				1			
	· · ·							763.7
	/Decrease th rilled rilling) Drilling) urs rking ng tage ement pply struction tation tation tai e Inserted Weterage (m) 24.00 32.00 117.00	08 07, 1995~ 11. 07, 1995~ 07. 10, 1995~ 07. 10, 1995~ 08 07, 1995 07. 10, 1995 05. 10, 1995 28. 11, 1995~ ength m 12r.ed 1100.00 /Decrease -90. 14 th rilling) 60.00 Drilling) 949.86 urs h 634.00 rking 538.50 ng 1753.50 2976.00 lage 40.00 ement 28.00 pply 612.00 struction 8.00 psituation 3594.00 e Inserted Weterage 07.111ng x100 (m) (11.00 11.7.00 11.7.00 11.7.20 1	11. 07, 1995~04 10, 1995 07. 10, 1995~27 11, 1995 10. 07, 1995 05. 10, 1995 28. 11, 1995~01, 12, 1995 ength m lared 1100, 00 /Decrease -90, 14 Core Length th rilling) 60, 00 Drilling) 949, 85 urs h 538, 50 19, 77 ng 1753, 50 58, 50 19, 77 ng 19, 73, 50 58, 50 19, 77 ng 1753, 50 58, 50 19, 77 10, 00 tation 3594, 00	08 07, 1995~09 07, 1995 11. 07, 1995~04 10, 1995 07. 10, 1995~07 11, 1995 10. 07, 1995 05. 10, 1995 05. 10, 1995 28. 11, 1995~01, 12, 1995 11. 00 ength m 12ned 1100, 00 Overburden 7. 00 rilling) 60, 00 Pecrease -90. 14 Core Length 924. 68 th rilling) 60, 00 Recovery 97. 35 0rilling) 949. 85 urs h \$38. 50 19. 77 15. 93 rg 1753. 50 58. 50 19. 77 15. 93 rg 1753. 50 58. 50 19. 77 15. 93 struction 8.00 0.01 <	08 07, 1995~09, 07, 1995 Dritting 11.07, 1995~04 10, 1995 139, 50 ing 10.07, 1995 05.10, 1995 139, 50 28.11, 1995~01.12, 1995 5, 50 12ned 1100, 00 Verburden 7, 00 //Decrease -90, 14 0.09, 86 Core Length 0.00, 000 Recovery //Decrease -90, 14 0.09, 86 Core 0.00 Recovery 97, 35 0, 00 - 100, 00 0.00 Recovery 97, 35 0, 00 - 100, 00 0.00 Recovery 97, 35 0, 00 - 500, 00 rilling 949, 85 0.00 20, 00 urs h % 100, 00 20, 00 rigg 588, 50 19, 77 15, 93 400, 00 100, 00 80, 60 0, 76 100, 00 100, 00 80, 00 100, 00 100, 00 100, 00	08 07.1935~09.07.1995 Drilling 84.00 11.07.1935~04 10.1995 Recovering 44.50 07.10.1935 139.50 Recovering 44.50 ing 10.07.1935 139.50 Recovering 44.50 ing 10.07.1935 139.50 139.50 Recovering 44.50 ing 10.07.1935 5.50 5.50 S.50 139.00 ength m n Core Recovery of 1.39.00 139.00 139.00 ength m n Core Recovery of 1.39.00 139.00 139.00 filled 1000.00 Dverburden 7.00 139.00 139.00 filling 60.00 Recovery 97.35 0.00-100.00 100.00 rilling 64.00 21.30 17.16 300.00-400.00 100.100 rking 588.50 19.77 15.93 400.00-500.00 100.56 rking 588.50 19.77 15.93 400.00-500.00 100.56 rking 580.00-90.00 <td>08 07.1955~09 07.1935 Drilling 84.00 5.00 07.10.1935~04 10.1995 139.50 Reovering 44.50 6.00 07.10.1935~02 11.1995 139.50 139.50 7.0</td> <td>08 07, 1955~09 07, 1995 Drilling 84, 00 5, 00 370, 00 ing 10, 07, 1935~27 11, 1095 139, 50 Recovering 44, 50 6, 00 190, 00 ing 10, 07, 1935~27 11, 1095 139, 50 139, 50 6, 00 190, 00 ing 10, 07, 1935 28, 11, 1995~01, 12, 1995 5, 50 5, 50 0, 00 21, 00 ength m m Core Recovery of 100m Hole 602, 00 ing 1100, 00 Dverburden 7, 00 132, 00 12, 00 602, 00 ing 1100, 00 Dverburden 7, 00 100, 60, 00 96, 45 95, 45 ing 100, 00, 186 Core (m) (N) (N) (N) rilling 60, 00 Recovery 97, 35 00, 00, 00 95, 45 95, 45 rilling 949, 85 19, 77 15, 30 000, 00, 100, 00 99, 77 95, 50 riking 588, 50 19, 72 15, 93 400, 00, 500, 00</td>	08 07.1955~09 07.1935 Drilling 84.00 5.00 07.10.1935~04 10.1995 139.50 Reovering 44.50 6.00 07.10.1935~02 11.1995 139.50 139.50 7.0	08 07, 1955~09 07, 1995 Drilling 84, 00 5, 00 370, 00 ing 10, 07, 1935~27 11, 1095 139, 50 Recovering 44, 50 6, 00 190, 00 ing 10, 07, 1935~27 11, 1095 139, 50 139, 50 6, 00 190, 00 ing 10, 07, 1935 28, 11, 1995~01, 12, 1995 5, 50 5, 50 0, 00 21, 00 ength m m Core Recovery of 100m Hole 602, 00 ing 1100, 00 Dverburden 7, 00 132, 00 12, 00 602, 00 ing 1100, 00 Dverburden 7, 00 100, 60, 00 96, 45 95, 45 ing 100, 00, 186 Core (m) (N) (N) (N) rilling 60, 00 Recovery 97, 35 00, 00, 00 95, 45 95, 45 rilling 949, 85 19, 77 15, 30 000, 00, 100, 00 99, 77 95, 50 riking 588, 50 19, 72 15, 93 400, 00, 500, 00

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Table 2-2-1 Summary of the Drilling Operation on MJZC-10

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Cals.	Drilling Lar.	th (a)		Daily Totel (#)		Shift (shift)		Kan Norking (aar	<u>г </u>
			Tote1 Qumulated	Orillina Lucgth	Core - Leigth	Dritting	Totat	Engineer	Barkar
3114	Rd-con	Bhift 2	9.00	9.00	0,00	0.00	1.00	1,00	1
5	Pda		0.00	8.00	0.00	0.00	1.00	\$.00 \$.00	10
i	Tra		0,00	D. 00	9,60		1.00	3.00	10
1	Raes	·	0.05 60.00	0.00 60.00	8.00 D.00	2.00	2.00	4.00	1
	24.00	36.00	60.00	0.00	0.00	0.00	1.00	2.00	2
1	Day off Reas	Day of t	50.00	8.00	0.00	8.00	1.00	4.60	!
	2.16		53. \$4	3.34	2. 37	2.00	2.00	<u>1.00</u> 1.09	
12	sla i n	1.10	59.44	6.10	: 6,10 18,40	<u>2.00</u> 2.00	2.00 2.00	4.00	
11	15. 50		87. 84 94. 18	18.49	5,97	2.00	2.00	4.00	
14	5.30		14 IS	8.00	0.00	1.00	1.00	3.00	
16	Exc-bt Day off	Day off Day off	94.10	\$. DO	0,00	9. DČ	0.00	2.00	
	2.5		96.75	2.57	2.57	1, 60	2.00		
10	Recov	Recov	96. 75	0.00	B. DO	B. D O	2.00	4.00	
19	Recev	Roccy	<u>\$6.75</u>	0.00	0.00	90.08 2.00	2.00		<u> i</u>
29	4.4		109.55	12.00	12, 49	1.00	2.00		
21	1.4		117.00	8.00	0.00	9.00	2.00	4.00	
22 23	Roza In-cp	Reas In-cp	117.00	0.00	8, 90	2.00	2.00		
24	1.3		123.83	6.83	6.06	2.00	2.00		
15	5.2	7 L 86	133.95	10. 13	9.66	2.00	2,00		
26	4.6		144.28	10.32	10.12	2.00	2,00		
17	5.1		154, 91 154, 91	0.005	1.00	6.00	0.00	2.00	
28	Day off	Duy off Day off	154.91	Q. 00	0.00	0.00	0.00		
	Day off	Day off	154.91	0,00	0.00	0.60	9.00		
31	7.6	5 5.82	168.59	13.68	13.58	2.00 1.00	2.00		
Mg 1	5.3		173.90	5, 31 D. 00	4, 88 8, 90	1.00	0.00		
1	Racov	Recov	173, 90 173, 90	D. 00	0.00	0.00	6, 0(4.0	
<u> </u>	ABCOV In-cp	Recov Ad-ex	173.90	D. 00	0.00	1.00	2.01	4.0	
5	Hain	Day off	175.40	0 .00	0.00	0.99	1.0		
	Day off	Day off	175.90		0.00	0.00	5.0		
1	10.0		194 65		34.85	2.00			
	17.3		229.50 254.40		23.56	2.00)
10	16.0		276.40		21.59		2.0		
11	12.1		303.20	25.00	26.37	2.00			
12 :	1.0		306.20	3.00	2.14				
13	Day off	Day off	306.20	8.00	0.00 5.90				
14	lard	9.70	\$15.50		0.94	2.00			
15	2.		314.05 324.40		3.21			0 4.0	0 1
<u>16</u> 17	10.5		351.90		26. 65	2.00			<u> </u>
18	- 13.		388. 4	14.50	16.30				
11	6.	12 2.88	\$75.70		I. 96 7. 30	2.00 1.00			
20		96 Day off	383.00		36.95				0 1
21			457,50		57.51	2.60	2.0		
21	13.		480.1	22.61	22. 61				
24	1.	9.50	493, 4		18.00				
25			521.7		22.25				
26			\$45.9 582.0		31.81				0 1
27.	17.		802.6		20.6	2.00			
21	Band	Recov	602.6	0.00	0.00				
30	ln ró	4, 62	607.5		474				
- 11			<u> </u>		32.3			10 4.0	0 1
Sep 1	H2.	43 17.50 Recav	663.9			6. O	0 2.0		
	Recov	Recov	663. 9	0.00	0.0	0 <u>0</u> .0	c 2.0		
4	Racov	Bacce	663.9	e d. D0	0.0				0
5	Bacev	Recov	663.9						
	Recov	Recov	563.9 563.9						iŭ l
1	Racov	Recov Tečse	563.9			0. D	0 1.1	0 40	0
	Cen	Dey off	663.9	0.00	6.0	a 8.0	0 1.0		
10	Gay off	Day off	663.9	0.00			0 B . ()0 <u>2.</u> (
11	Day off	Day Off	663.9			C D.D C 1.0			
12	la-rd	8-0(9.99)	663.9 663.9						NO
<u>1)</u> 11	R-D(2, 44) R-D(1, 45)	R-D(3, 32) R-D(7, 34)	563.9			0.5 0	¢ 2.1	10	N
15	B-D(12, 04)	Day off	663.9	0 0 0	9.0	0 1.0			
46	A-b(11.01)	A-D(9, 39)	663.1		0.0	0 2.0		10 4.1 00 4.1	
17	R-D(8.23)	A-D(9, 27)	663.1						
10	R-D(12.50)	R-D(5.50)	<u>663.</u> 663.		0.0	0 .0	0 1.1	00 8 .	00
<u>10</u> 20	ln ró	Day off Day off	663.		0.0	0.0	0 0.	00 1 .	
20	Dre-cee	One-cee	663. 5	N) B. D(0.0	0 0.0		00 4	
n	8-0(12.40)	A-D(14.40)	653.9				2.		
23	R-D(18,18)	R-2(3, 10)	65.5						
24	A-0(24.00)	Out-In-rd	653.	0 0.01					00
25	Cen	Day off Day off	663.			B. C	ю О.	2. 190	00
28	Day off Dra-cas	Day DTT Dre-cee	663.	0 Ø. Di	0. C	2.0	. 2	4	00
- 21	R-D(30, De)	14.5	\$76.	56 14.6	14	6 2.0	2.		
25	6	14 1n-rø	684.	70 5.7					
30		63 8.2							
Oct 1	12	.76 13.60 .00 12.0	5 <u>773.</u> 753.						Ó0

Table 2-2-2 Record of the Drilling Operation on MJZC-10 (1)

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Table 2-2-2 Record of the Drilling Operation on NJZC-10 (2)

		4	· .						1 A
·····	21.60	20.00	795. 70	41. 8 0	6.72	2.00]	2.00	1.00	10.0
	18.20	Day off	\$13.90	18.20	18.20	1.00	1,00	3.00	. 1.0
5	Day off	Day off	\$13.90	0. 00	R. 00	0.00	1.80	3, 90	9.0
	Day off	Bay off	613.90	0.00	D. 00	0.00	4.00	2.00	2.0
	Real	7.10	\$21.04	7.14	6. 60	1, 00	2.00	5.00	15.0
	12.00	14.00	\$51.04	30.00	29.90	2.00	2.00	1.00	10.0
	7.66	Racor	\$58.70	1. 66	7,51	1.00	2.00	1. 50	10.0
16	RECOV	Recor	\$58.70	1.00	F. 00	. 00	2.00	4,00	10.0
ii .	ia-Dut-rd	Exc-rd	\$58.70	1 .00	0.09	1.00	2,00	4.60	10.0
12	Tash	Fash	\$58.70	4.00	B. 00	4.00	2.00	L 00	\$0.0
: 13	lash	11.41	\$72.11	13, 41	13. ()	1.00	2.00	4, 80	30.0
14	24, 93	12.00	905.04	32. 93	\$2, 93	2. 00	2.00	L 00	10.0
15	11.06	8.55	924.65	23. 61	23.41	2.00	2.00	4.00)0.0
16	Recov	io-re	928.85	9,00	0.00	B. 00	2.00	. 00	
17	\$. 39	12.00	\$47.04	18.39	16.24	2.00	2.00	£ 00	
. 18	11.36	Racov	\$5\$, 40	11.36	10.63	1.00	2.00	4. b0	
18	Racov	PICOV	958, 40	4.00	ę co	F. 00	2.00	4.00	
20	Recov	Recov	\$58.40	. 00	0.00	. OD	2.00	4.00)0.(
21	Recov	Racer	958.40	8.00	0,00	B, 00	2.00	4.00	
12	Recov	Baccy	953, 40	1.00	0.00	0.00	2.90	5,00	
- 15	Racov	Rator	558.40	. 00	0.00	9.00	2.00	5.00	
-14	Day off	Day off	\$58, 49	9.00	0.00	0 .00	8,00	2.00	2
25	Racov	Racov	358, 40	1.00	0.00	0.00	2.00	5.00	
28	Racov	Racov	958, 40	8.00	0,00	8.00	2.00	5,00	10.
11	Racov	Recov	958, 40	D. 00	0.03	8,00	2.00	5.00	10.
20	Recov	Day off	958. 40	9.00	0.00	8,00	1.90	4.00	6.1
29	Recov	Racov	\$58.43	Ø. 00	0.03	0,00	2.00	5.00	10.
30	Pecov	Recov	958, 40	B. 00	B. DO	0.90	2.00	5.00	
\$1	Recov	Recov	958, 40	0.00	8,00	8.00	2.90	5.00	10.
Nev 1	Tex'ss	In rd	958, 40	0.09	0.00	1.00	2.00	5.00	18.
	R-D(8, 79)	R-D(9,70)	956.40	0.00	1. DC	2.00	2.00	5.00	10.
3	R-D(J, E1)	R-D(8.00)	958, 40	0,00	8, 50	2.00	2.00	5.00	18.
		A-D(5.00)	958, 40	0.00	8.00	2.00	2.00	5.00	10.0
5		R-D(8.59)	958. 40	0,00	\$, DO	2.00	2.00	5.00	10.0
		R-D(10.21)	958, 40	0,00	U . DO	2.00	2.00	5.00	10.0
— <u>;</u> —	R-D(15.01)	R-D(13.05)	958, 40	0.00	. B. DO	2.00	2.00	5.00	
· • •		R-D(11, 31)	956, 40	6, 80	9.00	2.00	2, 80	5,00	
		A-D (8, 35)	\$58, 40	0.00	8.00	2.00	2.00	5.00	
10	R-D(8.72)	R-D(13.56)	958, 40	0.00	e D0	1.00	2.00	5.00	
- 11	R-0(4.52)	In-co	958, 40	0,00	9,00	1.00	1.00	5.00	
12	Ince	\$JUTY	958, 40	0, 50	0. DO	B. DG	2.00	5.00	
13	Inred	Tash .	958, 40	0.00	0 DC	I. DO	2.00	5.00	
	8-2 (21, 55)	R-D(5. \$7)	958, 40	6, 60	0.03	2.00	2.00	5.00	
15	R-D (31. 32)	R-D (24, 39)	958, 40	8.00	8,00	2.00	2.00	5.00	
16	R-D(24.60)	R-D (3. 50)	- 958, 40	S . DO	9, 00	2.00	2.00	5.00	
11	R-0 (21 24)	R-D (27.00)	858.40	9 , DO	0.00	2 ,00	5.00	5, 40	
11	R-0(27.06)	R-D(17.06)	958.40	0. DO	2.00	2.00	2 . DO	5.00	
	R-D(22.67)	R-D(14 27)	958, 40	. BO	9.00	2.00	2.00	5.00	
21	R-D (19.19)	R-D(18.70	958.40	\$.00	0.00	2.00	2.00	5.00	
21	R-D(1.17)	R-D(4 40)	\$5.8, 40	1.0 0	0.09	2.00	2.00	5.00	10.
21	Recov	Recov	\$58.40	\$. DG	0 .00	0,08	2.00	5.00	
23	8-0(1.94)	R-D(18.05)	\$58, 40	9. DO	9. QG	2.00	2.00	5.00	
21	R-0(9.84)	R-0(14 17)	\$58.4G	1.00	0.00	2.00	2.00	5.00	
25	11.30	1\$.00	\$69. 70	\$1.30	91.30	2.00	2.00	5.00	
26	Out-In-ed	3.50	893.20	3.50	3.50	1.00	2.00	5.00	30.
27	1.70	B. 96	1009.06	18.66	16.15	2.00	2.00		
21	SUN	Quit-op	1009, 65	8.00	8.00	0.00	2.00	5.00	19.1
8	Diss	Cay off	1009.85	8.00	0.60	0.00	1.00	5.00	
50	Tra	Day off	1009.86	D. 09	9.00	0.00	1.00	5.00	
Doc 1	Disa	•	1009.85	B. 00	0. OG	B. 09	1.00	3. 80	
									1
	529.07	420, 79		1009.86	\$24.68	187.00	250.00	\$82.00	1330.

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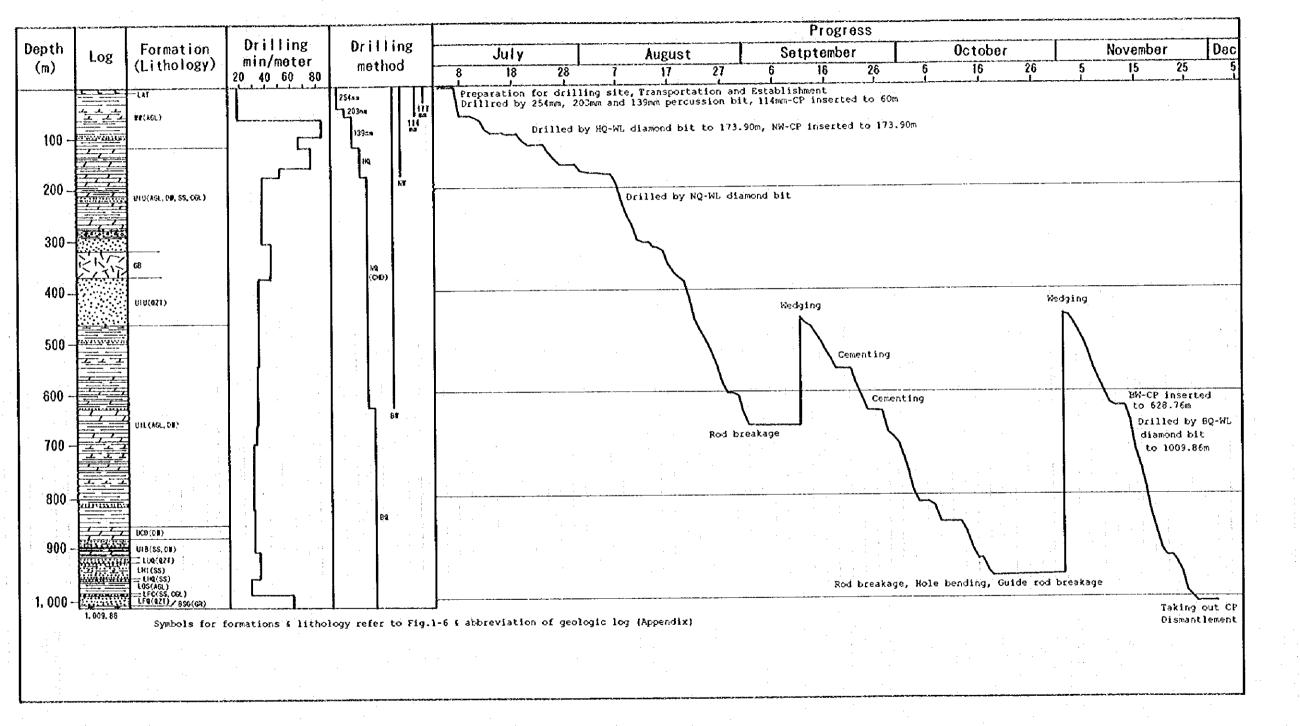
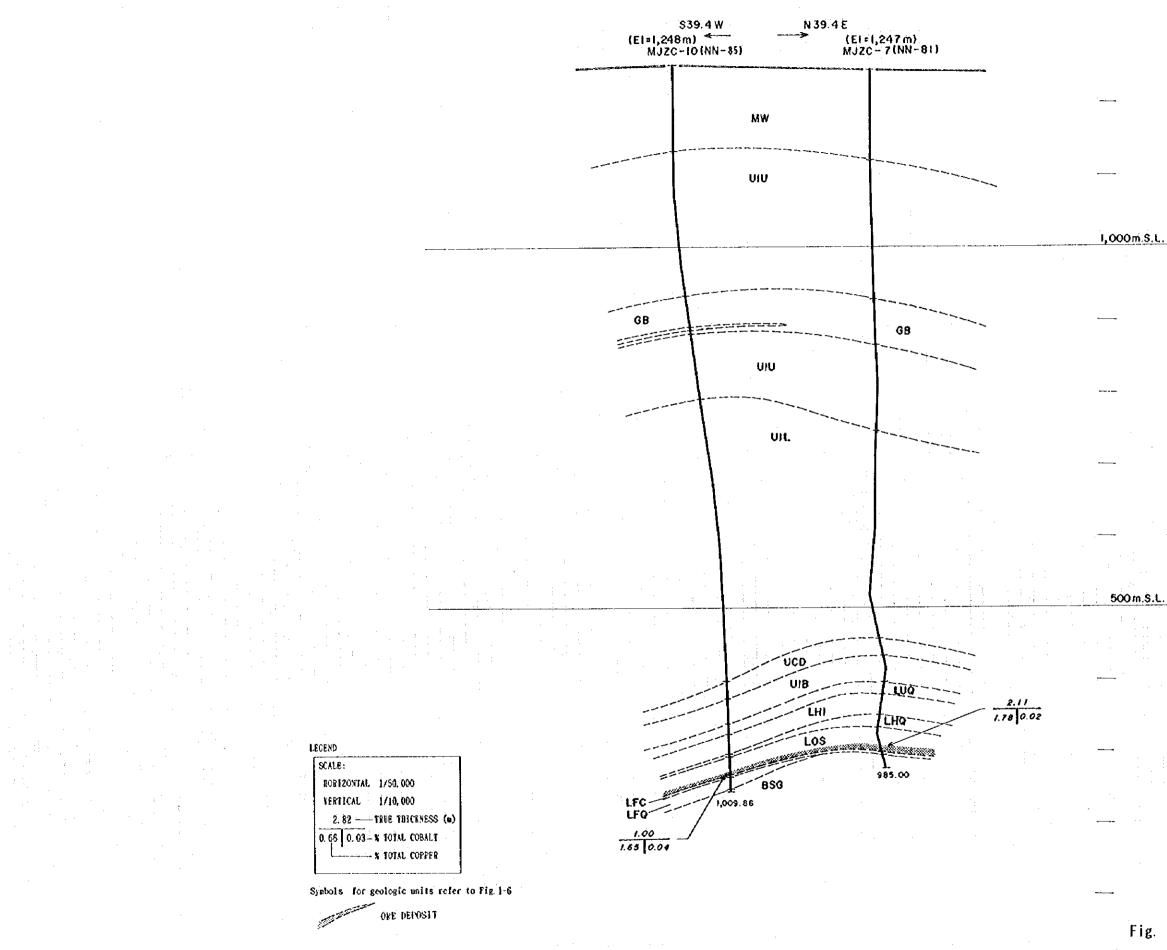


Fig. 2-2-1 Drilling Progress of MJZC-10

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1,000m.S.L.

Fig. 2-2-2 Geological Profile of Drill Hole (MJZC-10)

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drilled in the site of MJZC-9, and the tanker was transported to MJZC-10 by a tractor.

At 813.90m depth drilled, spindle bearing of drilling machine (Strata) was broken, and the machine was replaced by L-44 due to a long time to get exchange parts.

Borehole deviation was surveyed at 60m intervals. Surveying showed north-northeastward deviation (Appendices).

2-2 Geology and Mineralization

D

The geologic log is appended. The geology of this borehole compared to that of the survey area described in 3-2 of PART I, lacks the "Intermediate Conglomerate", "Feldspathic Quartzite and Grit" and "Basal Conglomerate" of the Footwall Formation, but otherwise it agrees well. Geological description of the borehole is as follows.

"Basement": 1,007.80 to 1,009.86m. The rock is altered coarse-crystalline granite. Strong biotitization and argillization of feldspar are observed in this rock. Also, weak dissemination of minute pyrite grains occurs.

Lower Roan Group

"Footwall Quartzite": 988.40 to 1,007.80m. It mainly consists of gray argillaceous quartzite and greywacke with pelitic parts irregularly. Anhydrite- biotite rock and thin bed of biotite rock occur in the basal part of this formation. Dissemination of chalcopyrite is observed at 988.40-995.20m.

"Footwall Conglomerate": 987.40 to 988.40m. It is composed of conglomerate and pebbly dolomitic sandstone. The conglomerate consists of quartz and feldspar granules. Pyrite dissemination is observed in this formation.

Shale Horizon": 961.30 to 987.40m. part The upper "Ore mainly consists of black sandy argillite with thin lamination. The lower part is made up of gray to black dolomitic argillite, and rich in biotite. The interval of 961.30 to 971.50m is the pyritized zone and 971.50 to 974.90m is pyrite-pyrrhotite zone and 974.90 to 979.80m is pyrite-pyrrhotite-chalcopyrite zone and 979.80 to 982.50m is pyrrhotite-chalcopyrite zone and

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982.50 to 987.40m is chalcopyrite-pyrite zone. The sulfides occur as dissemination at the bedding planes, the rim of dolomite concretions and patches or lenses of irregular shape. Results of ore assay are shown in Table 2-6-3. The cobalt mineral of this borehole was identified to be cobalt pentlandite (Table 2-5-2).

"Hangingwall Quartzite and Argillite": 957.90 to 961.30m. It mainly consists of siliceous greywacke, siliceous granule conglomerate, quartzite and pebbly argillite. Local weak dissemination of pyrite is observed.

(**1**.

"Interbedded Argillite and Quartzite": 927.70 to 957.90m. It is composed mainly of pelitic and dolomitic sandstone to greywacke with intercalation of thin dolomite, argillite and quartzitic lenses. Pyrite dissemination is observed in pelitic parts, and also, local anhydritization (lenses, patches) is observed.

"Upper Quartzite": 916.00 to 927.70m. It is composed of grayish white quartzite with pelitic bands. Pyrite dissemination is observed in the pelitic bands.

Upper Roan Group

"Interbedded Argillite, Dolomite and Quartzite": 879.00 to 916.00m. This is composed of thin-bedded alternation of pelitic sandstone, dolomite and argillite. The upper part is rich in sandstone, and the lower part is rich in dolomite and argillite.

Pyrite dissemination and silicification are locally observed in the basal part.

"Cherty Dolomite": 854.55 to 879.00m. It mainly consists of dolomite with anhydrite. In the upper part, dark green sandy argillite (Marker Shale) is intercalated. Weak dissemination of minute chalcopyrite-pyrite grains is observed between 869.70 and 878.00m.

"Arenite, Argillite and Dolomite with Anhydrite": 460.20 to 854.55m. The upper part mainly consists of dark gay thinly laminated argillite with small amount of thin local intercalation of dolomite and sandstone. The middle part mainly consists of an alternation zone of green to gray massive argillite and dolomite with small amount of thin quartzitic sandstone

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intercalation. The lower part is composed mainly of dark gray to black micaceous and sandy argillite with small amount of thin dolomite bed, thin sandstone bed, quartzite lens and grit. Contemporaneous brecciation and sheared zone are developed in above argillite, and many fractures are filled with dolothe or quartz or green clay. The middle and lower part have mite anhydritized (veinlets, patches, lenses) as a whole. been Pyrite dissemination is observed frequently in the middle part.

"Interbedded Argillite and Dolomite with Tectono-Breccias": 115.00 to 316.50m, 361.90 to 364.80m, 368.00 to 460.20m. The upper part mainly consists of an alternation of dolomite and argillite with thin intercalation of sandy rock and conglomerate. The conglomerate is composed of dolomite and argillite pebbles. The lower part mainly consists of quartzite with local pebbles of pelitic sandstone. Weak pyrite dissemination is observed frequently in the upper part.

"Mwashia Group": 7.00 to 115.00m. It mainly consists of gray micaceous to dolomitic argillite and black dolomitic shale with intercalation of dolomite and thin sandstone bed. Dissemination and lenses to laminas of pyrite-pyrrhotite are observed in the black shale.

"Gabbro (Amphibolite)" 316.50 to 361.90m, 364.80 to 368.00m. The bodies are dark gray altered and massive and strongly carbonatized and biotized.

2-3 Discussions

This borehole is located on a limb of a basement high (Figs. 2-2-2). It is inferred that this hole is located in the 1-7. vicinity of palaeo-basement high at the start of "Ore Shale" deposition. The mineralized zone of this hole is developed from the "Ore Shale" horizon toward the footwall, and zonal distribution of sulfide minerals are observed. The Fe/Cu ratio generally increases upward from chalcopyrite zones of "Footwall Quartzite" and "Footwall Conglomerate" to pyrite zone of upper This is interpreted to indicate that; this in "Ore Shale". area, the sea rapidly became deeper after the start of the deposition of the "Ore Shale" and consequently the environment became chemically reducing. Therefore, it is considered that the oreshoot was formed in this area because of the formation

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of a deep local depression in a relatively short period of time after the start of "Ore Shale" deposition on the limb of a palaeo-basement high. Ore shoots were not encountered in this hole, but the existence of rich ore in the vicinity is a possibility.

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3-1 Progress of Drilling

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The location and the collar elevation of MJZC-11 are appended.

Summary of the drilling, record of the drilling operation and the drilling progress are shown in Tables 2-3-1 and 2-3-2, and Figure 2-3-1, respectively.

the near surface zone to 36.00m, non-core drilling was For 254mm percussion bit, and 219mm casing pipes were made by to 33.00m. At 33.00 to 95.00m, non-core drilling was inserted pipes were 203mm percussion bit, and 141mm casing made by inserted to 95.00m. Percussion drilling was continued by 114mm bit to 126m where percussion was given up due to the increase of water flow out from near 32m depth. Then the method was to skid-mounted WL, after 114mm casing pipes were changed Cuttings were collected at 1 m intervals inserted to 95m. during non-core drilling.

WL Coring was done by HQ bit to 174.00m, and NW casing pipes were inserted to 174.00m, drilled further to 852.87m by NQ bit and CHD rods. At 557.15m depth drilled, the machine (Sullvin-22) was replaced by L-44 due to a small capacity of drilling.

Rod grease, cutting oil and Drillprops were used in order to prevent vibration during operation.

Waters were supplied after pumping up from a water borehole drilled in the site of this hole.

Borehole deviation was surveyed at 60m intervals. Surveying showed north-northeastward deviation (Appendices).

3-2 Geology and Mineralization

The geologic log is appended. The geology of this borehole compared to that of the survey area described in 3-2 of PART I, lacks the "Basal Conglomerate" of the Footwall Formation, but otherwise it agrees well. Geological description of the bore-

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		· ·	
Table 2-3-1	Summary of t	he Drilling Operation	on MJZC-11

			Surve	y Period		· • • • • • • • • • • • • • • • • • • •		Total Man	Day
		Perio	a	Day	Work Day		Off Day	Engineer	Worker
Operation									
Prepara	tion	23.07, 1995~	-24 07 1995					•••	
		28. 07, 1995							
		29.07.1995~	-06. 08, 1995						
		07. 08, 1995^	03.08.1995						
		15.09,1995		14, 5		5.5	9.0	35	
Drillin	g	25.07.1995~	-27. 07. 1995		-				
		09. 08, 1995^	13. 09, 1995		Drilling	42.5	4.0	164	3
		15.09.1995~	-25. 09, 1995	49.5	Recovering	3.0	0.0	12	
Dismanti	ling	14, 09, 1995						,	
		26.09.1995~	-04, 10, 1995	10.0		9.0	1.0	30	
Total				- 74.0	-	60.0		243	5
Drilling		n .		n	Core Recover	y of 100m 1	lole		
Length f			Overburden	6, 00				Core	
	e/Dacrease	-247. 13	Core Length	720, 63	Depth of Hole		Core	Recover	-
in Len		· · · · · · · · · · · · · · · · · · ·					Recovery	: Cumulat	əđ
Length I		852.87	-		(m)		(\$)	(1)	
	Drilling)	126.00	Recovery	99, 14		· · · · · · · · · · · · · · · · · · ·			
	Orilling)	726. 87	5		100.00-200.00		95.07	95.07	·····
Working He Drilling		h 491.00	49, 30	37, 61	200.00- 300.00 300.00- 400.00		98.55	97.07	
Other We	·	491.00	49, 30	31.71			100.00	98.14	·
Recoveri		91.00	9,14	6.97			100.00	98.92	
Subtote		996,00	100.00	76.29			98,86	98.91	
Reassem		46.00		3.52		<u>.</u>	100,00	99.07	
Dismant		30.00		2.30			100.00	99, 14	
Water Sc		189.50	na atan ƙanaria T	14.52		يدي. ا			
	nstruction	0.00	· · · · · · · · · · · · · · · · · · ·	0.00	·····				•• •. •. ·
Transpol	rtation	44.00		3.37	Efficiency o	fDrilling			
Grand To	otal	1305.50	·	100.00			m	day	#/day
Casing Pig	pe inserted				Drilling Period		852.87	49.5	17.23
		Kotorago	7		Total Length /	······································	m	shift	m/shift
Size	Veterege	Drilling	Length	Recovery	Totel Drilling		852. 87	68	12.54
		×10	0		Shifts				
	(m)	0)	(t)	Drilling Len	gth / Each	Bit (m)		
219mm	33.00	3. 87		0.0	8it Size	Drilled 1	ength	Cora Lan	gth
141mm	95.00	11.14		0.0	254 <i>m</i> m		36.00		, N,
114em	95.00	11.14		37.9	203aa		59.00		N
NV -	174.00	20.40		65.5	114mm		31.00		N,
8X	0.00	0. 00			HQ		48.00		44, :
	·			· · · · · · · · · · · · · · · · · · ·	NO		678. 87		676.
					80		0,00		0. (

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Date	Drilling Length	(1)		Jaily Total		Shift (shift	1)	Kan Jorking	(431)
	shife 1	shift 2	: Istal Dueufated	Drilling Length	Core Length	Drilling	Total	Engineer	Norke <i>r</i>
61 23	Pds	Day off	0.00	0.00	0.00	0,00	1,00	3.00	9,0
24	Reas	Dey off	0.00	9.00	0.00	0,00	1.50	3,00	9,0
25	68.00	Day off	65.00	65,00	0.00	1.09	1.00	3,00	7,0
75	29,00	Dey off	95.00	29.00	0.00	1.00	1,00	3.00	6.
27	31.00	Day off	125.00	31.00	8.00	1,05	1,00	3.00	6.
23	Dri-e/h	Day off	125.00	0,00	0,00	0.00	1.00	3.00	\$.
23	Day off	Dey all	125,00	D.00	0.00	0.00	0.00	2,00	2.
33	Day off	Day aff.	125.00	D_00	0.00	0.00	0.00	2.00	2.
31	Day off	Day off	125_00	D.00	0.00	0.00	0.00	2,00	2.
Asr 1	Day off	Day off	125.00	0.00	0.00	0.00	0.00	2.00	2.
2	Day off	Day off	125_00	0.00	D.00	9.00	0.00	2.00	2.
3	Day off	Day off	125.00	0.00	0.00	0.00	9.00	2.00	2.
•	Dev off	Dey off	125.00	D.00	9,00	D_00	0,00	2.00	2.
5	Dev off	Day off	125.00	0.00	0.00	0.60	0.00	2.00	2
5	Day off	Dey off	125.00	D. 00	0.00	0.00	0.00	2,00	2.
1	Tra-Reas-Joc	Dey off	125.00	0,00	0.00	0_00	1.00	3.00	13,
8	Reas	Day off	125_00	0.00	0.00	0.00	1.00	3,00	13.
8	Racov-co	Der off	125.00	0,00	0.00	0.00	1.00	3.00	13.
15	Recov-cp	Day off	125_00	0,00	0.00	0.00	1.00	3,00	13.
11	Racovicp	Dey off	125.00	0.00	9,00	0.00	1.00	3,00	13.
12	In-cp	Dey off	125.00	0.00	0.00	9.00	1,60	3.00	13. 2.
13	Der off	Dey off	125.00	0.00	0.90	0.00	0.00	2.00	2. 5.
[4	Reas-cp	Dey off	125.00	0.00	0.00	0.00	1.60	3.00	5. 5.
15		Recov-rd	130.50	4.50	4.28	1,00	1.00	3,30	B. 6.
35	1,59		132,09	1.59	1,50	1.00	1.00 2.00	1.00	
17	4, 13	3.21	139,40	7.31	7.25	2.00	2.00	.00	10.
38	6.75	5_08	151,23	11,83	13.83	2.00	2.00		13.
13	5.73	3,10	169,45	9,40	8,90	2.00	2.00		13
53	3,54	5_85		4,54	1,29	1,00	2,00		13
21	Haln .	0,50	174.00	0,00	0,00	9.00	2.00		10
22	In-cp	Ins-ce	174.00	9,02	9,00	1.00	2.00		10
23	In-rd	9.02	191,95	B.94	8.94	2.00	2.00		13.
24	7.37	1.57		29,33	20.25	2.00	2.00		10
25	4,53	15,80	212.29	30,00	30,03	2.00	2,00		10
25	15.60	14,40 14,13	252,42	20.13	20,13	2 13	2,00		10
27	5,00		278.29	15,87	14.97	2.00	2,00		10.
28	14.27		2/5.25	13,60	13.09	2.00	2,00		10
29	12.60		312,23	20,40	23,43	2.00	2.00		10.
30	8.00		329,29	8.00	8,00	1.00	1.00	are and a second second second	6
Sep 1	Day off	Day off	323, 29	0,00	0.00	0.00	D.00		- 2.
1	Day off	Day off	320.29	9.00	0.00	0,00	0.00		2
3	Day off	Day off	320, 29	And in the subscription of the subscription of the	9.00	3,00	0.00	2.00	2
1	17,23		350,29		30,00	2.09	2.00		13,
5	13,85		355,13	14,86	14,85	2.00	2.00	4,00	15,
5	12.93		431,15		35.00	2,00	2,00	4,00	10.
1	18.00		(41,35	40,20	40 20	2.00	2.00	4.00	10
8	19,83		457,15	25,80	25.80	2.09	2.00	4.00	10
9	Exc-ceble	6.00	473,15	5,00	5.00	1.00	2.0		10
10	Out-In-rd	12.00	485.15		12.00	2.00	2.60	4.00	10
- 11 -	17.00		514,00		28.85	2.00	2,00		10
12	Out-In-rd	9.35	523,35	9.35	9,35	2.03	2.00		10
: 13	15.80	18.00	557.15	33,80	33,60	2.00	2.00		13
14	Exc-Bac	Day off	557,15	0.00	0.00	0.00	1,00		
15	Reas	13,20	570,35		13,20	1,00	2.00		15
15	15.7	Day off	557.05	15,70	\$5,73	1.00	1,00	3.00	5
17	12.00		531.07	14.02	14,02	2.00	2.00	4.00	19
13	Recov-11	2,50	503,57	2.63	1,45	1.00	2.00	4,00	
19	19.45		\$41.05			2.00	2.00		13
50	20.30		580.35			2.00	2.00		13
21	20.7;						2,00		19
22		Exc-bt	741,75						
23	11,29		773.05				2.00		13
24	17.00						2,00		
25	21.9		852.87						
25	- Surv	Day alf	852.87						
27	Racov-cp	Day off	852.87						
28	Racov-cp	Day off	852.B7				1.6		
29	Recov-cp	Day off	852.87				1.0		9
30	Recov-co	Dey off	852.87				1.0		9
Oct I	Day Off	Der off	852.87				0.0		
2	Dise	Day off	852.87				1.0		
3	Tra	Day off	552 B?				1.0		
	Dise	Dev off	852.87	D_00	0.00	0.00	1,0	2.60	8

Table 2-3-2 Record of the Drilling Operation on MJZC-11

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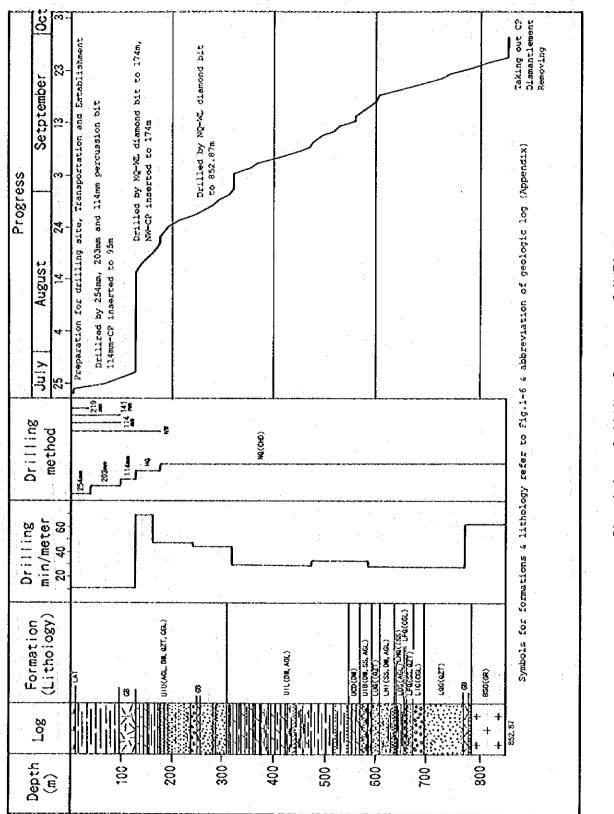


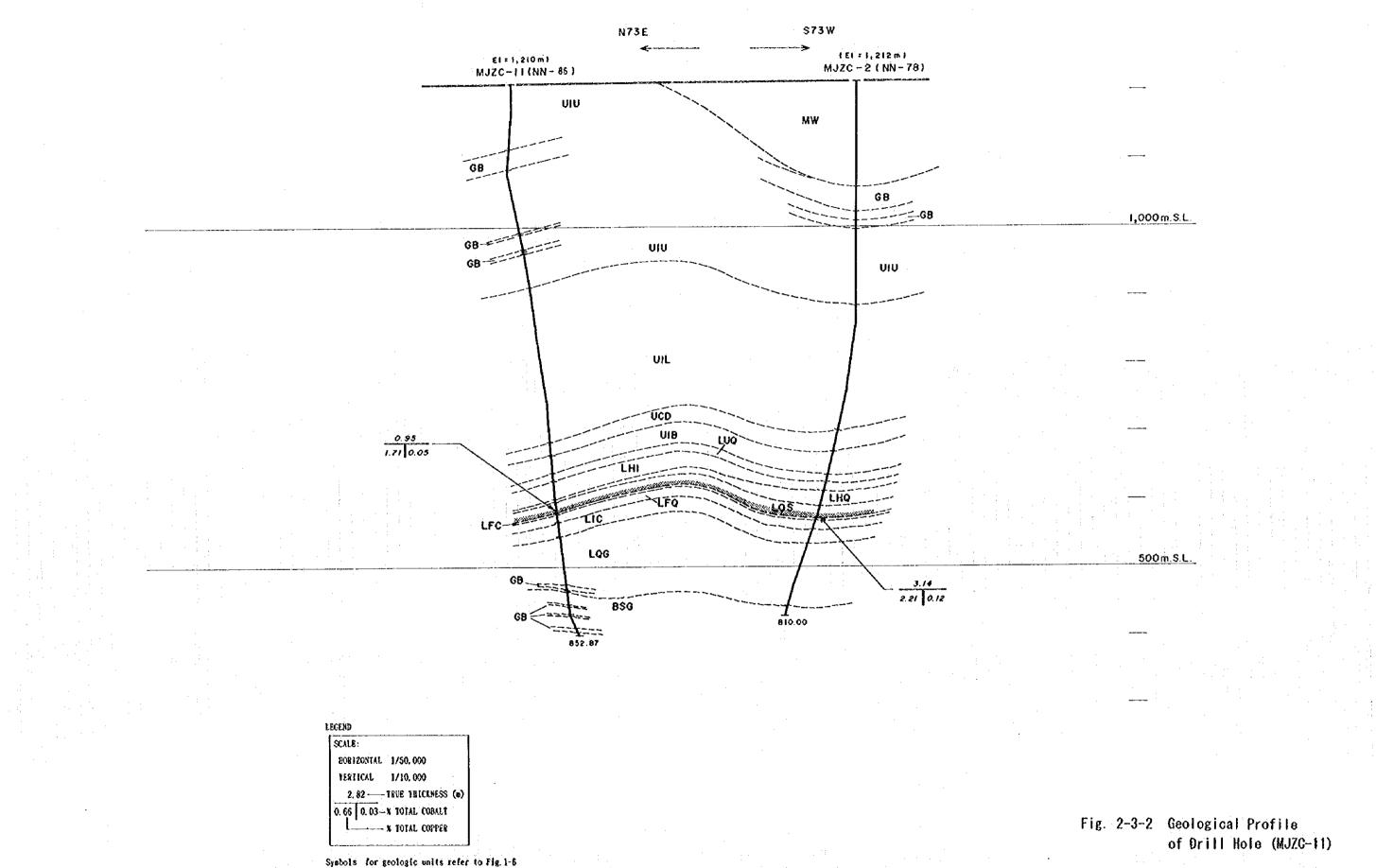
Fig. 2-3-1 Drilling Progress of MJZC-11

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OKE DEPOSIT

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hole is as follows.

"Basement": 785.70 to 852.87m. The rock is altered granite. Strong silicification or bleaching occur locally. Thin amphibolites are intercalated frequently in this rock. This amphibolite seems to be mafic intrusive rock.

Lower Roan Group

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"Feldspathic Quartzite and Grit": 697.00 to 771.60m, 782.10 to 785.70m. It mainly consists of white to gray quartzite. Pebbly quartzite and thin conglomerate bed are intercalated in the middle part. Thin beds of gritty and pelitic sandstone are intercalated in the lower part. The quartzite contains bands or dissemination of biotite. Also, amphibolite supposed to be mafic intrusive rock occurs in th lower part.

"Intermediate Conglomerate": 672.70 to 697.00m. It mainly consists of conglomerate with intercalation of small amount of thin quartzite bed in the lower part. The conglomerate contains various types of pebbles of rounded to subangular shape such as quartzite, sandstone, biotized argillite, chert, quartz, potash feldspar, granite and schist.

"Footwall Quartzite": 658.70 to 672.70m. It mainly consists of argillaceous to biotized quartzite and pelitic sandstone with many pelitic bands. Dissemination of chalcopyrite-bornite is observed in this formation.

"Footwall Conglomerate": 656.60 to 658.70m. It is composed of conglomerate with rounded to subangular pebbles. The nature of pebbles are quartzite, argillite, sandstone and small amount of granite.

"Ore Shale Horizon": 637.20 to 656.60m. It mainly consists of gray dolomitic argillite with intercalation of carbonaceous argillite in the lower part. Also, thin bed of dolomite and sandstone occurs in the basal part. The interval of 637.2 to 642.6m is the pyritized zone and 642.6 to 656.4m is pyritepyrrhotite-chalcopyrite zone. The sulfides occur as dissemination at the bedding planes, the rim of dolomite concretions, patches or lenses and veinlets. Results of ore assay are shown in Table 2-6-3. The cobalt mineral of this borehole was identified to be cobalt pentlandite (Table 2-5-2). "Hangingwall Quartzite and Argillite": 636.80 to 637.20m. It consists of white quartzitic sandstone with pelitic bands.

"Interbedded Argillite and Quartzite": 606.80 to 636.80m. The major rock of this unit is dolomitic or pelitic sandstone with many thin intercalations of dolomite and argillite. The sandstone is locally quartzitic. This formation contains patches to lenses of anhydrite.

"Upper Quartzite": 594.20 to 606.80m. This unit consists of pinkish gray quartzite with pelitic to micaceous bands.

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Upper Roan Group

"Interbedded Argillite, Dolomite and Quartzite": 566.70 to 594.20m. The upper part is composed mainly an alternation of dolomite and pelitic sandstone with intercalation of small amount of pelitic bands and thin quartzite beds. The lower part is composed of an alternation of dolomite and sandy argillite.

"Cherty Dolomite": 545.30 to 566.70m. The main component of this unit is white massive dolomite and green sandy argillite (Marker Shale) is intercalated in the upper part. This unit contains anhydrite patches. Also, minute chalcopyrite-pyrite grains is faintly disseminated in th lower part.

"Arenite, Argillite and Dolomite with Anhydrite": 306.70 to 545.30m. The upper part mainly consists of an alternation of micaceous dolomite and micaceous to dolomitic argillite with thin intercalation of sandstone to quartzite. The lower part is composed of argillite with grit. Anhydritization (patches, veinlets, lenses) occurs throughout the unit and limonitization by weathering is commonly observed in cavity of dolomite in the upper part.

"Interbedded Argillite and Dolomite with Tectono-Breccias": 6.00 to 96.00m, 126.00 to 222.70m, 224.50 to 247.60m, 254.30 to 306.70m. The upper part consists of micaceous or shaly or dolomitic argillite, limestone and dolomite. Lamina to dissemination is observed in a part of the shale. The major component of the lower part is quartzite with intercalation of dolomite and conglomeratic rocks in some localities. The conglomeratic rock consists of dolomite, quartzite and micaceous argillite pebbles of subrounded shape. Some gabbro bodies occur in this unit. Also, limonitization by weathering occur frequently in this unit, and limonite-dolomite-quartz veinlets which filled fractures of quartzite are considerably developed in the lower part.

"Gabbro (Amphibolite)" 96.00 to 126.00m, 222.70 to 224.50m, 247.60 to 254.30m. The bodies are dark green altered with platy cleavages developed and strongly carbonatized and biotized. Also, in the lower part of this hole, thin amphibolites similar to this rock are developed in the "Feldspathic Quartzite and Grit" of Lower Roan Group and basement granite.

3-3 Discussions

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The mineralized zone occurs in the "Ore Shale" in this borehole. The ores belong to the pyrite-pyrrhotite-chalcopyrite zone and pyrite zone. The location of this borehole was offshore far from the coast at the time of ore deposition and is inferred to have been unfavorable for copper precipitation.

Existence of a basement high is inferred to the west of this hole from the basement surface contour map (Fig. 1-10), but it is believed that this basement high was formed by folding after the deposition of the "Upper Roan Group" because all the beds of the "Lower and Upper Roan Groups" occur in harmony with the basement.

Chapter 4 MJZC -12

4-1 Progress of Drilling

The location and the collar elevation of MJZC-12 are appended.

Summary of the drilling, record of the drilling operation and the drilling progress are shown in Tables 2-3-1 and 2-3-2, and Figure 2-3-1, respectively.

For the near surface zone to 43.00m, non-core drilling was 254mm percussion bit, and 219mm casing pipes were made by to 34.00m. At 43.00 to 72.00m, non-core drilling inserted was percussion bit, and 177mm casing pipes made by 203mm were inserted to 48.00m. Percussion drilling was continued by 139mm bit to 180m where percussion was given up due to the increase of water flow out from near 67m depth. Then the method was changed to skid-mounted WL after 114mm casing pipes were inserted to 180m. Cuttings were collected at 1 m intervals during non-core drilling.

After NW casing pipes were inserted to 180.00m, WL Coring was done by NQ bit and CHD rods to 782.28m. At 550.51m depth drilled, the machine (Sullvin-22) was replaced by L-44 of MJZC-9 due to a small capacity of drilling.

Rod grease, cutting oil and Drillprops were used in order to prevent vibration during operation.

Waters were pumped up into a tanker from a water borehole drilled in the site of MJZC-11, and the tanker was transported to MJZC-12 by a tractor.

Borehole deviation was surveyed at 100m intervals. Surveying showed north-northwestward deviation (Appendices).

4-2 Geology and Mineralization

The geologic log is appended. The geology of this borehole compared to that of the survey area described in 3-2 of PART I, it agrees well. Geological description of the borehole is as

Table 2-4-1

Summary of the Drilling Operation on MJZC-12

	<u> </u>		Survey P	ariod				Total Fan	Cay
	-	Pario	(a) a property of the interpret of the interpret of the interpret of the interpret of the interpret of the interpret of the interpret of the interpret of the interpret of the interpret of the interpret of the interpret of th	Day	Work Day		Off Day	Engineer	Worker
		COLEG					-		
Operation	•	17. 07. 1995~18							
Preparation									
		23.07, 1995~25							
		17.09,1995~23	09,1995					31	
		23.09.1995		12				158	3
Drilling		19.07, 1995~22			Drilling		?		· · · · · · · · · · · · · · · · · · ·
		26.07.1995~15			Recovering		13	35	1 ÷
		24.09,1995~05	10,1995	68			F3		
Dismantlin	;	15.09.1995							
pranore re-	•	06. 10, 1995~10	10 1995	- 6		5		18	<u></u>
Total				86		58	28	242	4
			t		Core Recovery	of 100a He	10		
Orilling Lon			Guarburdan	29.00				Core	
Longth Pla					Death of Hole		Gore	Recove	er y
Increase/D		-317.72	COLD FOURTH	J 34. VJ	Ventil el trola		Recovery	Currula	tec
in Length		رجابا كمكت برادي					(6)	(1)	
Lorigth Del	11ed	782. 28			<u>(m)</u>				
OV/C Dri	illing)	180.00	Recovery	\$8, 63			99.78	99.78	
(Core Dr	(filing)	502.28			100.00-200.00			89 46	
Rorking Hour	1	h ·	1	<u> </u>	200.00- 300.00		89.39	98.77	
Dritting	I	340,00	47.16		300.00-400.00		\$7.94		
Other Work	ling .	313.00	43, 41		400.00- 500.00		97, 36	98.33	
Recovering		68,00	9, 43	7.33	500.00- 600.00		100.00	98.73	
		721.00		77.69	600.00- 700.00		99,50		
Subtotal		24.00			700.00- 800.00		\$8.32	98.63)
Reassemble		18.00		1.94					
Dispantler		111.00		11, 96					
Bater Sup		12.00		1.25					
Read Cons				4.53		Drilling			
Transport		42.00			Totel Length /		N	day	m/da
Grand Tot.		\$28.00	L	100.00	Dritting Period		282.28		
Casing Pipa	Inserted	ويحص والجاري والمعاد			Total Length /			shift	s/shi
		Esterage/	_				782.28	•	
\$ìze	Kuteraga	Drilling	ength	Recovery	Total Drilling		1 102.00	1 ~	ĭ) [ĭ
		× 100			Shirts		1		
	(m)	(6)		<u> (%)</u>	Drilling Long	th / Each 8	HT (N)	Core L	and the
219676	34.00	4,35		0.0		Drilled		1	engta
17765	48.00	6, 14		0.0	25400		43.00		
114cm	180.00			0.0	203mm		29.00		
<u></u>	190 00			76.6	139:00	1	108. C		
<u></u>	0.00				HQ		0.00		0
					NQ	1	602.2	· · · · ·	594
l		·			89	- 1	0.00	N	0

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Ca's	Dolling Ling	(h (s)		Daily Yotal (=)		Shife (shife)		Has Burking (san	}
(total		Core	0.11	1.1.1	Engineer	
1111	53.92 F	shift 2 Bay off	Curutated 8.00	tarath 0.00	Length 0.00	Britters 0.00	fotel \$.00	3.00	Rorker
- <u></u>	Rd-con Tra-Rest	Cay off	0.N	0.00	0.00	e .00).00	1.00	11.0
19	72.00	Cay off	12 60	72.00	9 .00	1,00	1.0(3.00	7.0
20	47.00	Cay off	119.00	47.00	() ,(v)	1.00	1.00	3.00	6.0
21	41.60	Cay off	360.00	47.00	0.09	1.00	1.00	3.00	0.0
22	20.00	Pay off	160 061	29.00	0.00	1.00	.00	3.00	6.0
23	fea Pess	Cuy off	130.06	0 .00	0 00	0.00	1.60	3.00	<u> </u>
	Reas	Cay off	160.06	9.00	0.00	0.00	<u> </u>	3.00	
	In-cp	Dir off	160.0C	23 83	0,00 22.89		<u>).00</u>	3.00	
26	23.63 10.94	Cay off Day off	213.9	10 24	10.94	1,00	2.00	3.00	5.0
28	Cay of I	Par off	2 3.9	0.pc	0.00	6.00	9.00	2.00	2.0
20	Ber off	Day off	213.9	3 DC	0.00	0.00	9.00	7.00	2 (
30	Cay off	Cay off	213.97	0.00	2 .00	0.00	g_00	2.00	2.0
31	30.01	Cay off	224.05	10.00	10.05	1.00	1,60	3.00	
Ave I	24 00	Day off	249 03	24.00	23.95	1,00	1.00	3 00	6.(
2	21.00	Cay off	269_03	21.00	20.74	1.00	1.00	3.00	5.0
	20.45	Car off	269.43	20.45	20 16 0 00	1.0€ 6.00	1.00	3.00 3.00	50
	Nain Dau att	Cay off	269.45	3.D(0.00	9.00	0.00	2.00	1.0
	Day off	Day off Pay off	289.4	J ,00	0.00	0.00	0.00	2.00	
	Dey off	Cay off	289_48	0.0(0.04	0.00	0.90	2.00	2 0
	Day off	Cay off	289.48	0.64	0.00	0.04	9.00	2.00	2.0
	Dey off	Cay off	2\$9.49	9.00	0.60	0.00	9.0ú	2.00	2.0
10	Cay off	Cay off	299.45	0 OC	0.00	0.00	0.00	2 00	2.0
<u>n</u>	Cay off	Cay off	283.48	9.00	0.00	0.00	0.00	2.00	2.0
12	Na n	Cay off	289.49	<u> </u>	0.60	0.00 0.00	0.00	3.00	8.0
- 13	Eay alf 0.55	Day off	299.03	9.55	9.55	1.00	1.60	\$.00	6.
14	12.00	Day off Day off	317.03	12.00	12.00	\$ 00	1.00	3.00	8.0
- 18	\$ 96	Pay off	317.01	5.9(5 96	F.00	1.00	3.00	6.0
17	12.02	Day off	\$29.03	12.07	(),60	1.00	1.00	3.00	6.0
18	24.00	Say off	353.03	24.00	21.00	- 1.0 0	1.00	3.00	6.6
19	16.88	Cay off	\$79.8	18.85	17.23	F.00	1.00	3.80	6.0
20	20.12		392.03	20.12	(9.17	1.00	1.00	3.00	
21	1.6	Cay off	400.43		8.0	<u> </u>	1.00	3.00	ļ <u>ę.</u>
21	12.10	Cay off	412.53	12 10	12.10	1.00	1.00	3_00 3_00	0.0 6.0
23 74	15.50	Cay off Cay off	4 (9.03	13.00	13.50	t.00	1.00	3 00	
25	1,58	Cay off	10.6	150	1.39	1.00	1.00	3.00	6.0
20	Out-In-ed	Day off	44).Et	0.00	0.00	0.00	1.00	3.00	0.0
27	19.42	Cey att	451.03	19.42	16.97	F 90	1.00	3.00	8.(
24	0.20	Pay off	470.51	9.28	9.26	00.1	1,00	3.09	8.0
29	Cay off	Cay off	470.5	0.00	0.00	0.90	0.00	2.00	5.0
30	Day off	Day off	470.51	0.00	0.00		0.00	7.00	20
31	Par off	Day off	470.51	0.00	0.00		<u>0.0(</u> 9.0(2.00	50
Sep 1	Day of t	Cay off	470.31	0.00 0.00	0.00		6.00	2.00	2.0
	Day off Day off	Cay off	470.31	0.00	0.00		0.00	2.00	2.0
4	Pay off	Cuy off	470.3	0.90	0.00	0.00	0.01	2.00	2.0
6	far off	Oay off	470,31	0.00	0.00	0.00	9.00	1 00	3.0
	Car off	Day uti	470.31	0.00	0.00	9.00	0.00	2.00	2.0
		Day off	+73.03	1.17	2.77	1.00	1.00	3.00	6.5
	1.00	Cay off	479.03	<u> </u>	6.00	<u>+.00</u>	1.00	3_00	0.0
10	12.00	Pay off	491.03	12.00	12.00	00.1 100.1	<u> </u>	3.00	8.0
	8 00	Day off	509.03		6.00		1.00	3.00	6,0
12	15 25	Pay off	\$24.28	15.25	15 25	7.00	1.00	3.00	6.0
13	1.65		532.83	8.55	8.55	9.00	1.50		6,0
1	12 22		16.6	12 22	12 82	1,00	1.00	3.00	6.0
15	5.45	Day off	\$59.51	5.45	5.46	3.00	1,00		\$,0
1	D-se-Fra	Day off	\$50.51	0.00			1.00		<u>. 11.0</u>
<u> 1</u>	Day off	Day of (\$50.51	0.00	0.00	0.00	0.00	2.00	2.0
	Day of	Cay off	\$50.51	0.00	0.00 0.00		0.00 0.00	2.00	2.0
20	Cay off	Pay off Pay off	550.51	0.00	0.00		0.00	2 00	2,0
21	far off	Cay off	550.51	0.00	0.00 0.00	0.00	0.00	7.00	2.0
27	Cay off	Day off	\$50.51	9.00	1 .00		0.00	2.00	
23	ly n	Our off	\$50 51	0.00	0.00	9.00	1.00	3.00	0.0
24	D.t-In-rd	20.47	\$70.93	20.42	20.42	2.00	2.06	4.00	10.1
8	21.27	6.09		10.11	27,77	2 <u>00</u> 2 00	2.0- 2.0-	4.00	
24	10.76	1.00	619.48	\$7.75	37,76	2.00	3.04	4.00	10.0
- 22	10 24	15.16		33,40	33,40		<u>} % % % % % % % % % % % % % % % % % % %</u>		10.0
	20.52	14.64		35 23	33,73 21-81	2.00	200		10.0
39	19.2 ² 12.06	Cay off	710.70		21.00			4 00 3 DG	10.0
211	Day off	Cay off	710.76	0.06	4.00	1.00 1.00	0.6	2.00	2 (
	22 21	1,79			23.00		2.04	4.00	10 (
	92.DC	3,50			14 54		2.00	4.00	.10,1
1	9,53	4.79	768.70	8.32	8.32	2.00	2.04	4.00	19.0
5	15.58	Şur # 1	762 28	15.58	15,40	1.00	2.00	4.00	10.0
• 6	Surv	Day off	782.26	9.00	0,0 0	0.00	2.00	4.00	U.U.
1	Øise-Tea	Pay off	782.28	0.00	0.00		1.00	4.00	0
- <u>+</u>	Pey off	Cuy off	782 28		0.00		0.0		
	10	Quy off	787.28	0.00	0.00	G.D0	1.00	3.00	<u> </u>
10	D: se	ł	782.28	9.00	0.60	0_0 0	1.0K	2 00	<u> </u>
	712.46	69.82			592.05				

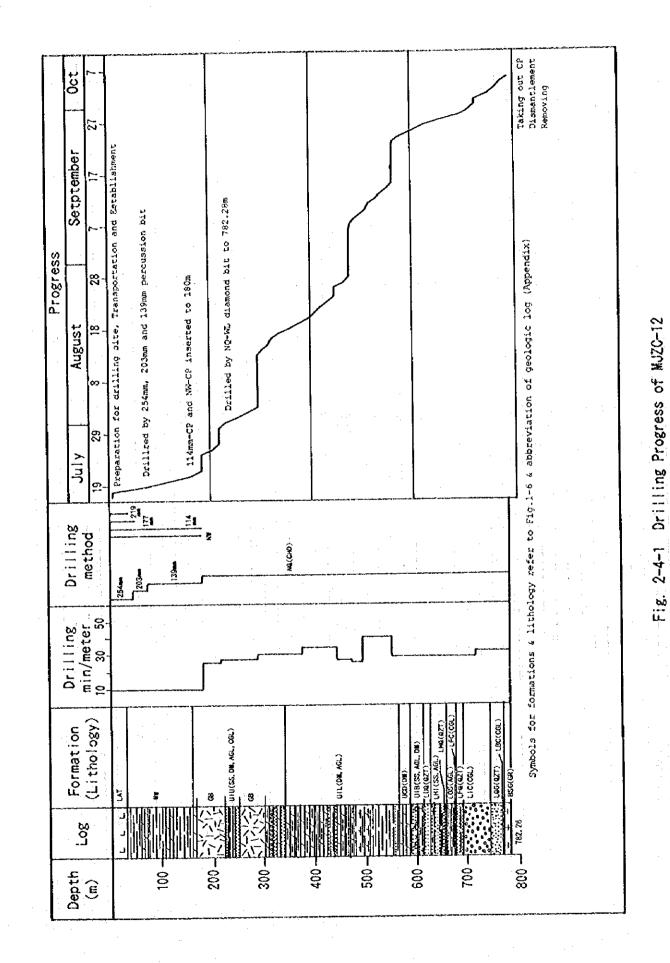
Table 2-4-2 Record of the Drilling Operation on MJZC-12

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follows.

"Basement": 769.60 to 782.28m. The rock is coarse-crystalline granite and rich in secondary quartz and biotite. Strong silicification and a large amount of mica occur in the top part. Thin mafic rocks occur in this rock.

Lower Roan Group

"Basal Conglomerate": 768.70 to 769.60m. Conglomerate with boulders of granite and fragments of potash feldspar.

"Feldspathic Quartzite and Grit": 743.70 to 768.70m. It is composed of pinkish grey quartzite with local pebbly quartzite, and contains bands or patches of biotite.

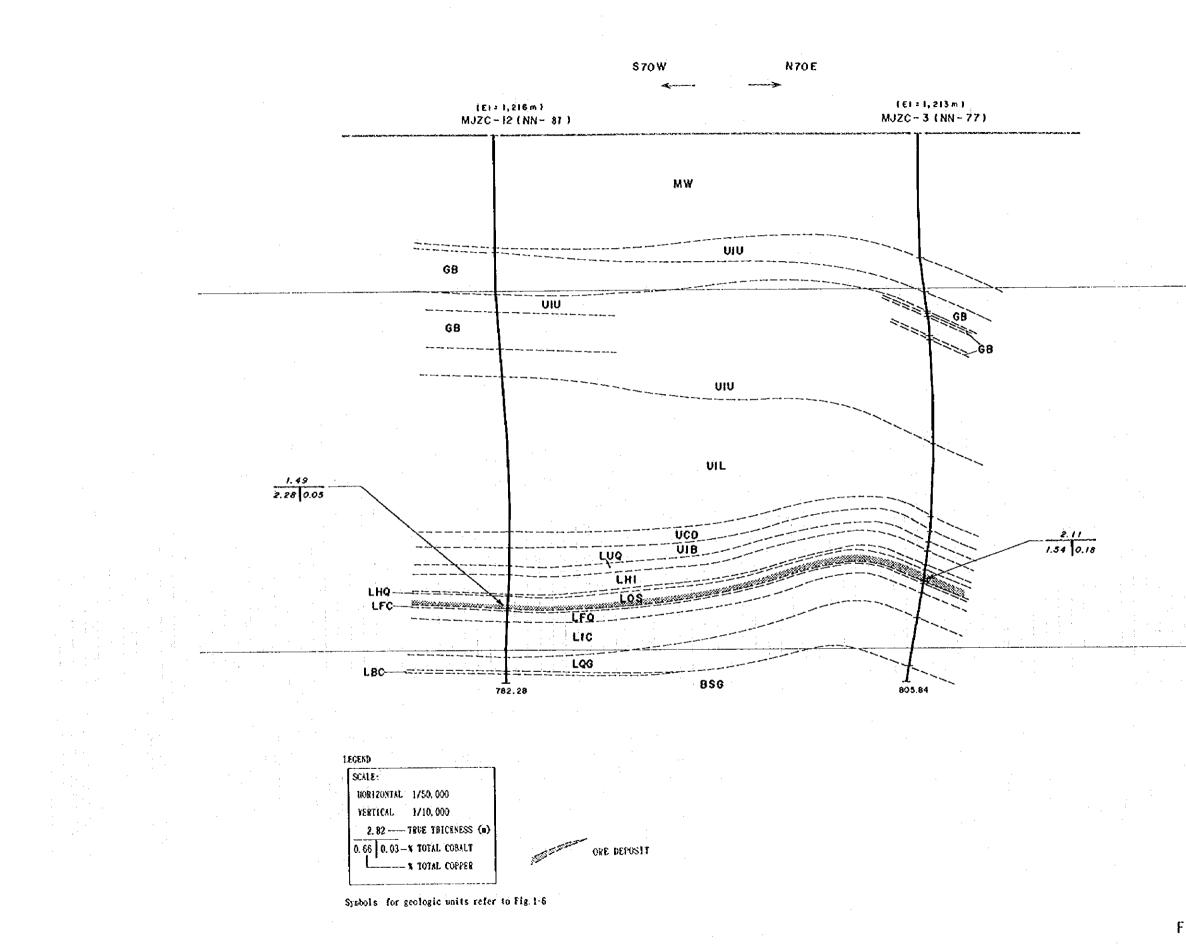
"Intermediate Conglomerate": 692.00 to 743.70m. It mainly consists of conglomerate with subangular pebbles, and contains locally granule conglomerate or pebbly quartzite. The nature of pebbles are rich in quartzite, chert, argillite, granite, quartz, schist and potash feldspar in the upper part, and rich in biotite schist, spotted mica schist and quartzite in the lower part.

"Footwall Quartzite": 675.50 to 692.00m. It mainly consists of pebbly sandstone, pebbly quartzite and pelitic sandstone, and many pelitic bands are intercalated in a state of thinbedded alternation in the sandstone.

"Footwall Conglomerate": 674.10 to 675.50m. It is composed of conglomerate with subangular to rounded pebbles. The most abundant pebbles are granite, followed by quartzite and biotite schist.

Shale Horizon": 655.40 to 674.10m. It mainly consists "Ore of black to gray sandy argillite with local dolomitic argillite. Also, thin bed of pelitic sandstone occurs in the basal part. The interval of 655.4 to 661.4m is the pyritized zone and 672.1m is pyrite-pyrrhotite-chalcopyrite zone and 661.4 to 674.0m is bornite-chalcopyrite zone. The sulfides 672.1 to as dissemination to lamination at the bedding planes , occur patches to lenses of irregular shape. Results of ore assay are shown in Table 2-6-3. The cobalt minerals of this borehole were identified to be cobalt pentlandite and carrollite (Table 2-5-2).

E



1,000 m.S.L.

500 m.S.L.

Fig. 2-4-2 Geological Profile of Drill Hole (MJZC-12)

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"Hangingwall Quartzite and Argillite": 654.00 to 655.40m. It consists of pink to white quartzite with pelitic bands.

"Interbedded Argillite and Quartzite": 624.00 to 654.00m. It is composed mainly of an alternation of pelitic sandstone, quartzite and argillite, and thin beds of dolomite are intercalated in this unit. Anhydrite lenses occur commonly in this unit.

"Upper Quartzite": 612.20 to 624.00m. This unit consists of white to pink quartzite with pelitic and micaceous bands.

Upper Roan Group

"Interbedded Argillite, Dolomite and Quartzite": 584.70 to 612.20m. This unit is composed of thin-bedded alternation of dolomite, argillite, pelitic sandstone and quartzite.

"Cherty Dolomite": 562.70 to 584.70m. The main component of this unit is white massive dolomite with anhydrite. Micaceous and dolomitic argillite (Marker Shale) is intercalated in the upper part.

"Arenite, Argillite and Dolomite with Anhydrite": 339.00 to 562.70m. The upper part mainly consists of an alternation of dolomite and greenish grey argillite, and contains locally intercalations of pelitic sandstone. The lower part is composed mainly of green sandy to gritty argillite with thin intercalations of small amount of dolomite and sandstone. Anhydritization (patches, veinlets, lenses) occurs throughout the unit and limonitization by weathering is observed in the upper part.

"Interbedded Argillite and Dolomite with Tectono-Breccias": 157.00 to 164.50m, 220.50 to 247.60m, 300.20 to 339.00m. The part consists of an alternation of dolomite and argilupper lite, and contains intercalations of sandstone. The lower part is composed mainly of an alternation of quartzite and dolomite, and contains intercalation of conglomerate. The conglomerate consists of pebbles of dolomite and argillite. Some gabbro bodies occur in this unit. Also, limonitization by weathering occur generally this unit.

"Mwashia Group": 29.00 to 157.00m. It is composed mainly of dark grey to black shaly argillite with intercalation of dolomite. Dissemination of pyrite is observed locally in the argillite.

"Gabbro (Amphibolite)" 164.50 to 220.50m, 247.60 to 300.20m. The bodies are dark green to black altered, massive. Generally, strong biotitization and carbonatization occur and silicification is observed locally.

4-3 Discussions

The mineralized zone occurs in the "Ore Shale" of this borehole and vertical zoning of sulfide minerals is observed. "Ore Shale", bornite-chalcopyrite zone occurs In the at the bottom and the mineral assemblage changes upwards through pyrite-pyrrhotite-chalcopyrite zone to pyrite zone. This fact indicates that this site was located in the vicinity of the palaeo-basement high at the beginning of "Ore Shale" deposi-But subsequently the sea rapidly became deeper and tion. the environment became chemically reducing and thus unfavorable for copper precipitation.

Although this borehole is inferred to be located over а basement high from the basement surface contour map {Fig. 1the units of the "Lower and Upper Roan" Groups occur 10), in harmony with the basement complex. Thus this basement high is considered to have formed by folding after the deposition of the "Upper Roan Group". The occurrence of palaeo-basement high in this basement high became a possibility as above mentioned. If, therefore, deep depressions were formed locally on the limb of the palaeo-basement high during the beginning of "Ore Shale" deposition, ore shoots could have been formed.

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Table 2-5-1 Results of Microscopic Observation of Thin Sections

	lexture		granular	granular	schistose	granular	granular	granuiar	schistose	chistose	granular	granuiar	
		Ch Zr Op All others	<u></u>	E.	Š	12	18	12	\$(Ba (A) schistose	2	2	
		ALL (•	•	4	•			•	·	
		8	⊲	4				•	4	Þ	•	•	
		Zr		•						•	•		
		ਲੂ ਤ	4		4		•		0			•	
	Crystal Fragment	Ap {Ep	_						· .				
		aA i			_								
) 0		•	⊅						•	•	
_ :	Cryst	Rut To											
ryst/		NC 5		A	0	0	0	0		0		0	
Phenocryst/			8	0	0	0	4	0	0	0	0	•	
		Anh Bi	0	0	0				0		0		
		පි		0					-				
-		١d				0	© ©	ŧ		-	0	00	
		0z Kf		0	0	©	<u> </u>	╊╼	┟┯╸	6	Ô		
				-	-			F					
	Rock Name		anh rock	meta-ss	Bi schist	metagranite	1.1.1	meta-ss	Bi-Ch rock	mica schist	meta-ss	metagranite	
	Forma	t G	1-1	LFO	50		BSG					BSG	
	locality		1132.50	1144 00	1007.30	827 00 BSG	852 80	701 20	723 50	732-00	750 00	782.00	
	003	Depth (m)	NJZO-9.	NJZC-9.	NJ70-10	N.17C-11	N.170-11	N.170-17	N. 170-12	N.170-12	N.17C-12	NUZC-12	
	Samle	No	106	902	1001	T	100	-1	1 202	1203	1204	1205	

Abbreviations

Abundance of minerals: \odot : abundant, \bigcirc : common, Δ : a few, \cdot : trace

Rock : ss:Sandstone,

Mineral : Oz:Quartz, Kf:Alkali feldspar, P1:Plagioclase, Ca:Carbonate, Anh:Anhydrite, B1:Biotite, Mc:Muscovite, Rut:Rutile, To:Tourmaline, T1:Titanite, Ap:Apatite, Ep:Epidote, Ch:Chlorite, ZriZircon, Op:Opaque minerals All:Allanite, Ba:Barite

Results of Microscopic Observation of Polished Thin Sections(1) Table 2-5-2

													·						1	·							
sample	Looality, F	Formar	Desaription				Ō	Ore kineral	eral			•					Gang	97.	Gangue Minera!								
Ż		tion	:	8	۲ ۲	Po E) 8	3	S	8	SP ot	others	8	Χf	0 1d	Ca 81	*	Rut	2	Υ.	8	£	ઈ	₽ N	A11	others	
P-901	NJZC-9, 1087 90 LOS	ရှိ	Co dise in col Arg	0	0						(•)eg	:	0	0	4	0	0 0				·				4	(•) Эд	-) holi.
P-902	WJZC-9 1109 40 1	Ş	Go bleb rich in dol Arg	0		4 1	Ocat						0		0	0 0	0	ŀ			Ŀ						
P-903	#J20-9 1111 90 LOS	105	Co betch rich in soh Arg	0	A	÷			•				Ô			Ľ	0 0	•	ļ				٩				
p-904	#UZC-9, 1112 40 LFO	20	Co diss in nic Se	4	-		Ocat	ا ا	•				0	4	Ĕ	0	0		ŀ			· _				Xen (-)	
					<u></u>		- car																				
P-905	NJZC-9 1114 20 LFO	<u>9</u>	Co diss in pebbly CCT	0				:	· · ·		т. Тп.(0	0	4	0	0	•	•	•		_		ŀ			
P-906	MJZC-9, 1120, 00 LFO	с Б	Co. Co dize in Q7	0		Ļ	0	•	•				0	4	0	0	♦	Ŀ			-	•		•			
P-1001	P-1001 MUZC-10, 965 50 L05	587	Py dies in Arg	-	0				-		표(•)	Ŷ	ŀ	Ő	0	0	<u>0</u>	ŀ						1	1=	(-) uo <u>w</u>	
P-1002	MUZC-10. 974.60 LOS	8	Py-Po diss in bodding plane	4	0	0	- pent	۲					4	Ť	0	0	0		ļ					1-			
		:	and rim of dol spot				· .		<u>.</u>	<u> </u>				:	•							:					
P-1003	P-1003 MJ2C-10 983 30 L05		Co-Po irr lone in Arg	0	0	4	\$ 8	- co pent	•	1.	1		0	-	Ĕ	0	0		•		•			•			
P-1004	P-1004 #JZC-10, 936.60 [05		Co-Py diss in Arg	0	Ö	÷	8	· co pent	•				0	י ק	4	0	0		ŀ							-	
P-1005	P-1005 #JZC-10, 992 40 LFO	1	Co-dise in CZ1	0					•		(-) Ba (-)		0	0	v ▼	0	0	Ļ			•	-	ŀ	•			
P-1101	WUZC-11 643 30 1.05		Py-PoXcp dise in dol Arg	•	0	Ö	8 0	Oco pent				 - -	0		4	0	o	•		•				 			
P-1102	P-1102 MUZC-11, 648 00 LOS	1	Po-Co-Py spot in Arg	0		0	- 00	- co pent	•	•	(•)4T		0		0 0	4 0	▼.▼	•						·			
P-1103	MJZC-11, 651.30 LOS		Co-Py-Po-do! lens/spot/vit	6	0						•		©	·		⊲ √	0 4						٩				
			in carb Arg				:			÷		:								, ,		_					:
P-1104	P-1104 MUZC-11 656.30 LOS		PyXQD-Po diss in dol-arg Se	0	7	٩	. 60	- co pent			Ura	Ura(•)	0	0 O	~	0	6	•									
P-1201	P-1201 MJZC-12, 661 50 (LOS		Co-Py-Po-dol diss in Arg	0	0	0	8	co pent	•		.		0	-	0	0	0		•			•.					
P-1202	P-1202 MJ20-12, 664.00 LOS		Op-Py-Po-dol tamina/spot	0	9 0	4							0	: •	H	0	0	•		:		•		•		(-) uoji	
P-1203	#UZC-12, 670, 50 LOS		Coirr tens in Arg	0	0	Ø	8.	co pent					0		Ľ	0	0		•				0				
P-1204	1,20-12, 671 10 L	SOJ	P-1204 WizC-12, 671, 10 LOS - 02-02-02-02 Ions in Arg	0	0	0	98	co pent	•				0			م	0	•	•			1	ŧ				
P-1205	P-1205 MJC-12, 673.30 LCS	8	Bo patch/lens/lam & diss	0	•	-	👁 - car						0	Ĕ-	0	⊲ 0	0				•						
		;]	in mic Se	{		-				<u>·</u>						: 1						:					
Abbreviations	lions																										
Abundance	of minerals: 4	48 : Ø	Abundance of minerals: ${f O}$: abundant, ${f O}$: common, Δ : a few,	for,	•	trace		•			2 																

Rook : dol:dolomitic. Arg:argiliite. 027:quartzite. ss:sandstone.schischistose. mic:micaceous, carb:carbonaceous

Wineral: Bo:Bornite, Co:Chalocoprite, Po:Pyrrhotite, Py:Pyrite, GniGalena, BS:Mative Simmuth, SP:Schalerite, dissidissemination CatiGattionite, Car:Carrollite, Pen:Pentiandite, Co pen:Cobelt pentlandite, Be:Berlee, Th:Thorite, Ura:Uraninite, Dz:Cuertz, Mf:Mali Hoidcaner, Py:Pyrise;Colares, Ga:Carbonate, Ph:MyMPyrite, Bi:Biotite, Mc:Maccovite, Rat:Battia, To:Tournamine, T1:Titanine, Do:Labonate, Mh:MyNPyrite, Bi:Biotite, Mc:Maccovite, All:Allanite, Mon:Monazite, Xen:Xenotime, REEREConte, Vitveinlet S.

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Results of Microscopic Observation of Polished Thin Sections(2) Table 2-5-2

COMPOSI LI OTIS (M LA)	(MCD)		ł	ľ				7.4.2
mineral	S	Fe	D	3	5	Z	AS	10131
cattierite	50.32	13. 32	0.07	35.09	8	0 2 2 3	8 8	_ 99. 21
cattierite	46. 75	21.59	00	26. 21	8	8	4.48	<u> </u>
carrollite	39.26	2.08	14. 32	44, 38	8	0. 13	0.01	100.17
co pentlandi	30.94	2.52	0.0	63. 98	0.0	5	8	1 66
so pentiandi	30.86	3.50	0	64.07	0.14	1.54	8	100.12
carrollite	40.24	0.02	14. 63	4	0.8	0, 37	0.00	99 74

Atomic ratio								
No. Imineral	Ś	Ee.	Cu	So	Zn	Ni	AS	Total
P-902 cattierite	65, 08	9.89	0.05	24. 69	0.0	0. 28	0 8	8 8 8
P-904 cattierite	62.05	16.45	8	18.93	8	0.03	2.55	100
P-904 carrollite	54.60	1.66	10, 051	33.58	0°	0. 10	8	8
P-1101 co pentiandi	45.31	2.11	8	50.97	0.0	1.60	0.0	8
P-1204 co pentiandi	4.95	2.93		50.79	0.10	1. 23	0.0	8
P-1205 carrollite	55.86	0.02	10.25	33.60	0,00	0.28	0.00	100.00
				Analysis	s by E	PMA of	JEOL.	-

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Table 2-5-3 Results of Chemical Analysis of Ore Samples (1)

MJZC-9

Sumple	1 1	T-Cu	AS-Cu	1-Co
No KC12701	(+) 1077.45 ~ 1077.95	(1)	(1)	(<u>1)</u> (0,01
KC12702	1077.95 ~ 1078.45	(9,01	<0.01	<0,01
KC12703	1078,45 ~ 1979.00	<9.01	<3.01	0,01
KC12704		<9.01	<0.01	0,01
KC12705 KC12705	1079,50 ~ 1079,67 1079,57 ~ 1080,17	0.05 9.85	<0.01 <0.01	<0.01 <0.01
KC12757	1080,17 ~ 1080,17	0.09	<0.01	<3_01
KC12708	1080,87 ~ 1081.17	0.05	<0.01	<0.01
KC12709	1081,17 ~ 1081,67	0.05	<9.01	<9.01
¥C12719	1081.67 ~ 1082.17	0.12	<0.01	<0.01
KC12711 KC12712	1082,17 ~ 1082,67 1082,67 ~ 1083,17	0.17	<0.01 <0.01	0.01 <0.01
#C12713	1063.17 ~ 1083.67	0.15	(0.0)	0.01
KC12714	1083.67 - 1084.27	9,22	<0.01	<9.01
KC12715	1084.27 ~ 1084,68	0,32	<0.01	0.01
XC12715 KC12717	1064.68 ~ 1085.18 1065.18 ~ 1085.68	0,45	<0.01	0.01
1012715	1085.68 - 1085.18	1.45	0.0	0.01
KC12719	1066,18 - 1065,68	2.44	0.01	0.01
KC12720	1065,68 2, 1087,18	2,58	0.01	0.01
KC12721	1087.18 - 1087.75	2,59	0.01	0.01
KC12722	1087,75 ~ 1068,25 = 1068,25 ~ 1088,75	2,35	<0.01 <0.01	0.01
EC12724	1065.75 ~ 1089.26	0,70	<0.01	0.01
1012725	1069.25 ~ 1089.50	1,53	20.01	<0.01
IC+2725	1089.50 ~ 1090.00	0.02	<0.01	0.02
RC12727 RC12728	$\frac{1090.07}{1090.50} \sim 1090.50$	0.13	<0.51	0.02
NC12729	1091.00 ~ 1091.50	0.69	0,01	0.03
KC12730	1097.50 ~ 1092.00	D, 33	(0,01	0.02
RC12731	1092.00 ~ 1092.50	0,03	<0.01	0.03
KC12732 KC12733	1092,50 ~ 1093,00 1093,00 ~ 1093,50	0.02 0.01	<0.01 <0.01	0.02
XC12734	1093,50 ~ 1094,00	0.02	<0.01	0,02
¥C12735	1094.00 ~ 1094.54	D_02	<0.03	0.02
*C12738	1094,50 ~ 1095,00	0.01	<0.01	0.02
KC12737 KC12738	1095.00 ~ 1095.35 1095.35 ~ 1095.85	0.00 0.20	<0.01	0.02 0.03
1012735	1095.85 ~ 1095.35	0.23	<0.01 <0.01	0.03
KC12743	1095.35 ~ 1095.65	0.21	<9.01	0.01
KC12741	1095,85 ~ 1097,35	0.25	<9.01	0.01
KC12742 KC12743	1097.35 ~ 1097.85 1097.85 ~ 1098.35	0.25	<0.01 <0.01	0_02
1012744	1098.35 2 1098.85	0.02	<3.01	0.02
KC12745	1098.85 ~ 1099.35	0_27	<3.01	0.02
KC12745	1099.35 ~ 1099.85	0.05	<3.01	0.03
XC12747 XC12745	1099,85 ~ 1100,35 1100,35 ~ 1400,85	0.39	<9.01	0.02
KC12749	1100.85 ~ 1101.35	0.01	<0.01 <0.01	0.01
KC12750	1101.35 ~ 1101.85	0.01	<0.01	0,02
KC12751	1101.85 ~ 1102.35	0,01	<0.01	0.02
KC12752	1102.35 ~ 1102.85	0.01	<0.01	0.03
KC12753 KC12754	1102,85 ~ 1103,35 1103,35 ~ 1103,85	0.01	<0.01 <0.01	0.03
KC 12755	1:03.15 ~ 11:04.35	0.01	<0.01	0.05
KC 12758	1104.35 ~ 1104,85	0.41	<0.01	0.05
RC12757	1104.85 ~ 1105.35	0.08	(0.01	0.09
KC12758 KC12759	1105.35 ~ 1105.76 1105.75 ~ 1105.25	0.06 0.07	<0.01 <0.01	0.07 <0.01
4012753	1136.28 ~ 1105.76	0.07	<0.01	0.05
¥512761	1106.75 ~ 1107.26	0.05	<0,03	0,05
#C12762	1107.25 ~ 1107.75	0.11	(0.01	0.64
KC12763 KC12764	1107.76 ~ 1108.16 1108.15 ~ 1108.35	0,03 0,63	<0.01 <0.01	0.04 0.05
KEILINS	1108.36 2 1108.85	2.55	(9.0)	0.62
KC11768	1104.66 ~ 1109.35	1.99	· (0, 01	0.01.
RC12767	1109.36 2 1109.85	2.88	(J.01	0.02
EC12754	1103.55 2 1110.35	1,99	0.01 1	0.02
KC12753 KC12770	1110.35 ~ 1110.85 1110.65 ~ 1111.35	3.73	0.01	0.07
KC12771	1111.36 2 1111.75	1.178	Q.01	0.01
reizin	1111,76 ~ 1117,28 2	9.59	9.01	0.25
1012773	1112,28 2 1112,76	1.88	0.01	0.91
IC11224	100,75 ~ 100,85 100,75 ~ 100,75	1.97	0.01 (0.01	0.64
1012775	1119.18 ~ 1116.28	1.38	0.01	0.02
KC12777	1111.26 ~ 1114.76	0.12	<0.01	0.61
KC12775	1111,26 ~ 1115,26	0,87	(),01	0.02
KC12779 KC12780	1115.25 ~ 1115.76 1115.76 ~ 1116.26	0.75 0.79	0.01 0.01	0 01
1 2/50	1113.78 ~ 1119.00	4.13	~J.VI	

Sasple	Depth	T-Cu	AS-Cu	I-Co
Xo	(1)	(1)	(1)	(1)
KC12781	1115.25 ~ 1115.75	9.38	(0.01	<0.01
KC12782	1115.75 ~ 1117.00	0.45	<0.01	<0.01
KC12783	1117.60 ~ 1117.50	1.06	0.01	<0.01
KC12754	1117.50 ~ 1118.00	3,47	0,01	<0.0t
KC12785	1113.00 ~ 1118.50	9.62	<3.01	0.01
KC12785	1118,50 ~ 1119.00	9,33	<3.01	0.01
KC12797	1119.00 ~ 1119.50	9.77	<0.01	(0,0)
1012785	1119.50 - 1120.00	1.93	0.01	0.08
KC12789	1120.00 ~ 1120.50	0.73	<0.01	0.03
KC12793	1120,50 1120,75	0,32	<0.01	0.01
KC12791	1120.75 ~ 1121.25	0.41	<0.01	0.01
KC12792	1121.25 2 1121.75	0.19	<0.01	<0.01
KC12793	1121,76 ~ 1122,26	0.04	(0.0)	<0.01
KC12794	1122.26 ~ 1122.75	0.04	<0.01	0.01
KC12795	1122,75 ~ 1123,25	0.02	<0.01	<0.01
KC12795	1123,25 ~ 1123,75	0.07	<0.01	<9.01
KC12797	1123,75 ~ 1124,26	0.03	(0.0)	0.01
KC12798	1124.25 ~ 1124.75	0,01	(0,0)	0.01
KC12799	1124,75 ~ 1125,26	0.01	(0.0)	0.01
KC12800	1125,25 ~ 1125,76	0.03	<0.01	0.01
KC15101	1125.75 ~ 1125.25	9,01	<0.01	0.01
KC15102	1126.25 ~ 1126.75	0.02	<0.01	0.05

Fidth	Depth	Ĩ-Cu	AS-Cu	T-Co
<u>()</u>	(1)	(1)	(\mathbf{x})	(D)
2.58	1055.68 ~ 1088.26	2.29	<0.01	<0.01
5,90	1108.35 ~ 1114.26	3.12	<9.01	0.08
5.26	1114.26 ~ 1129.50	D.78	<0.01	<d.02< td=""></d.02<>

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Table 2-5-3 Results of Chemical Analysis of Ore Samples (2)

MJZC-10

MJZ0-11

	-10 Depth	T-Cu	Ás-Cu	[-Co
Sasple Xo	Depth (=)	(1) (1)	(1)	(1)
1012222	953.70 ~ 954.70	0.01	<0.61	0.03
1012223	954.70 ~ 955.70	<0.01	<0.01 <0.01	<0.01
KC12224 KC12225	955.70 ~ 956.70 958.70 ~ 957.70	0,01 <0,01	<0.01 <0.01	<0.01 <0.01
KC12225	958.70 ~ 957.70 957.70 ~ 958.70	<0.01	0.01	<0.01
KC12227	958.73 ~ 953.73	<9.01	<0.01	<3.01
KC12228	953,70 ~ 953,70	<0.01	<0.01	<9.01
KC12229 KC12229	$\frac{950,70}{951,33} \sim \frac{951,33}{951,33}$	<0.01 <0.01	<0,01 <0,01	<0.01 (0.01
KC+2231	951.33 ~ 951.83 951.83 ~ 352.33	<0.01	<d.01< td=""><td>(0.0)</td></d.01<>	(0.0)
KC12232	952.33 ~ 952.79	<0.01	(0,01	<0.01
KC12233	952.70 ~ 953.20	<3.01	<0.01	<0.01
KC12234 KC12235	953.20 ~ 953.70 953.70 ~ 954.20	0.01 <0.01	<0.01 <0.01	<0.01 <0.01
#C12236	951,70 ~ 954,70 954,20 ~ 954,70	0.01	<0.01	<0.01
KC12232	954,70 ~ 965,20	0,03	<0.01	0.01
KE12238	955,23 ~ 955,73	0.01	<0.01	0.02
KC12239 KC12243	955.70 ~ 955.20 955.20 ~ 955.79	(0,01 (0,01	<0_01 <0.01	0.01
KC12241	965.23 ~ 955.73 965.73 ~ 957.23	<0.01	<0.01	9,01
KC12242	957.20 ~ 957.73	<0.01	<0_01	0.01
KC12243	957.70 ~ 958.20	0.01	<0.01	10.01
KC12244 KC12245	958,20 ~ 958,70 958,70 ~ 959,20	<0.01 0.01	<0.01 <0.01	<0.01 0.01
KC12245	958.70 ~ 959.20 959.20 ~ 959.70	(0.01	49.01	<0.01
KC12247	959.73 ~ 970.20	<0_01	<0.01	(0,01
KC12248	970.23 ~ 970.70	0.07	<0.01	<0.01
KC12249 KC12250	970.70 ~ 971.23 971.20 ~ 971.70	0.01 <3.01	<3.01 <3.01	<0.01
KC12251	\$71.70 ~ \$72.20	0.01	<3,01	0.01
KC12252	972.20 ~ 972.70	<0.01	<0.01	0.01
KC12253	972.70 ~ 973.20	0.01	<0.01 <0.01	0.01
KC12254 KC12255	973.23 ~ 973.62 973.62 ~ 974.12	0.02	<0.01	0.03
KC12256	\$74.12 ~ 974.62	0.03	<0.01	0.03
KC12257	974.62 ~ 975.12	0.18	<0.01	0.03
KC12258 KC12259	975.12 ~ 975.52 975.62 ~ 975.15	0.22	<0.01 <0.01	0.04 0.02
KC12260	975.62 ~ 975.15 975.15 ~ 975.55	0.11	<9.01	0.02
KC12261	975.55 ~ 977.15	0.23	<0.01	0.02
CC12252	977.15 ~ 977.55	0.32	<0.01 <0.01	(0.0)
KC12253 KC12254	977.55 ~ 978.16 978.15 ~ 978.55	0.12	<0.01	0.01
KC12255	978.65 ~ 979.16	0.12	(0,01	0.01
KC12255	979.16 ~ 973.66	0.05	(D.01	0.01
KC12257 KC12253	979.66 ~ 983.15 950.16 ~ 953.70	0.06	<0.01 <0.01	0.01 0.01
AC12269	380.16 ~ 983.70 983.70 ~ 961.20	0.06	<9.01	0.01
KC12270	951,20 ~ 981,73	0.05	<9_01	9.01
4012271	981.70 ~ 982.20	0.11	(9.01	0.02
KC12272 KC12273	982, 20 ~ 592, 20 582, 70 ~ 583, 20	1.52	<0.01 (0.01	0.03
KC12274	983,20 ~ 983,70	0.83	<0.01	0.03
KC 12275	983,70 ~ 984,20	0.23	(0.01	0.04
KC12275	984,20 ~ 984,70 984,70 ~ 985,20	0.01	<u>(0,01</u> (0,01	0.03
KC12278 KC12278	984.73 ~ 985.23 985.23 ~ 985.73	0.15	<0.01	0.04
KC12279	965.70 ~ 985.20	0.05	<9.01	0.07
KC12280	985.20 ~ 985.72	0.79	<0.01	0.03
KC12281 KC12282	985.70 ~ 987.23 987.20 ~ 987.58	0.11 0.34	<0.01 <0.01	0.02
KC12203	987.20 ~ 987.58 987.55 ~ 938.06	0.29	D.01	(9.0)
KC12284	988,05 ~ 988,55	0.54	0.01	<0.01
KC12285	938,55 ~ 989,06	0.41	0.01	<9,01
KC12285 KC12287	999.05 ~ 989.70 999.70 ~ 990.20	0.24	<0.01 <0.01	<0.01 <0.01
KC12289	999,70 ~ 990,20 990,20 ~ 990,70	0 39	(0.01	(0,01
KC12289	993.20 ~ 991.20	0.25	0.01	<0_01
KC12293	991.23 ~ 591.88	0.05	<0.01 <0.01	<0.01
KC12291 KC12292	991.88 ~ 992.38 992.38 ~ 992.88	0.25	0.01	<0.01
KC+2293	992.35 ~ 992.68 992.88 ~ 993.56	0,11	0.03	(3.0)
RC12294	993.55 ~ 994.05	0.34	5.01	<0.01
KC42295	994.06 ~ 994.55	0_32	<9.01	<0.01
KC12295 KC12297	998.55 ~ 995.06 995.08 ~ 995.55	9,35	<3_01	<0.01
KC12298	995.58 ~ 995.06	0.15	0.08	<0.01
#C12299	995.06 ~ 995.55	0,10	0.01	<0.01
KC12330	995.56 ~ 997.05	0.09	0.82	<0.01 <0.01
KC13902	997.05 ~ 397.55	1 9,03	1 (0.01	
	Depth	ks-Cu	1-00	٦ -
¥idth (•)	(•)	(O)	(\$)	

Sa	aplė (Depth	T-Cu	AS-Eu	T-Co
	Ь	(1)	(1)	(1)	(D)
KCI	6103	635.05 - 635.55	<0.01	(0.01	().01
KCI	5104	635.55 ~ 636.05	<0.01	<0.01	0.01
KCI	\$105	635,05 ~ 635,55	<0.01	<0.01	0.01
KCI	\$105	838.55 ~ 637.05	<0.01	<0.01	<0.01
KCI	\$137	\$37.05 ~ \$37.55	<0.01	<9.01	0.01
KCI	5108	\$37,55 - 638,05	<0.01	<9,01	0.02
IC.	\$109	638,05 ~ 638,55	<0.01	<9.01	0.05
111	5110	\$38.55 - \$39.05	<3.01	<0.01	0,03
XCI	5111	\$39,05 ~ \$39,55	<3.01	<0_01	0.02
KCI	5112	839.55 ~ 643.05	<0.01	<0.01	0.02
KCI	\$113	543.05 ~ 643.55	<0,01	<0.61	0,02
RCI	5114	640,55 ~ 641.05	0.01	<3.01	0,02
KCI	\$115	641,05 ~ 641,55	<0.01	<0.01	0.02
(C)	5116	641,55 ~ 642.05	<0.01	<9,01	0.02
108	5517	542.05 ~ 542.55	<0.01	<0.01	0.03
XC	5118	642.55 ~ 642.75	<0.01	<0.01	D.02
KC1	5719	642.75 ~ 643.25	0,02	<0.01	0.05
10	5)23	643,25 ~ 643,75	0.02	(0.0)	0.14
	5121	643.75 ~ 644.25	0.91	<3.01	0.07
4C	6122	544,25 ~ 644,75	0_01	<3.01	D.D3
(C)	6123	644,75 ~ 645,25	0.02	<0.01	0.02
¥C.	6124	545.25 ~ 545.75	8.01	<0.01	0.02
KC 1	6125	645,75 ~ 645,25	0_05	(0,01	D.D2
	6125	545,25 ~ 545,25	0.09	<0.01	0.02
	16127	645.75 ~ E47.05	0.15	×0.01	0.02
	6728	647.05 ~ 647.55	0.28	<0.01	0.02
	6129	647.55 ~ 649.05	9.53	<0.01	0.02
	6130	649.05 ~ 643.55	0.25	<0.01	0.01
	15131	645.55 ~ 649.05	0.21	(0.81	0.01
and the second sec	15132	649.05 ~ 643.55	0,33	0.01	0.01
	\$133	643.55 ~ 650.05	0.21	<0.01	0.02
	15134	650.05 ~ 650.55	0.27	: <3.01	0.01
	\$135	\$50,55 ~ \$51,05	3.43	0.01	0,02
	6135	\$51.05 ~ \$51.55	9.0	D_01	0.02
	6137	651,55 ~ 652.05	0.13	0.01	0.01
	15138	\$52,05 ~ 652,55 \$52,55 ~ 653,05	0.13 0.24	(0.01 (0.01	0.01
	15139	652,55 ~ 653,05 653,05 ~ 653,35	0.17	<0.01	0.02
	15143	553, 35 ~ 653,85	0.22	<2.01	0_02
	15141	653,85 ~ 654,35	0.44	0.01	0.02
	13142	654,33 ~ 654.85	1.15	0.01	0.04
	18192. 18199	654.85 ~ 655.30	1.55	<0.01	0.06
	16145	655.30 ~ 655.80	0 57	<0.01	0.07
	16145	555,83 ~ 656,33	0.24	10.01	0.07
	15147	555.30 ~ 856.70	0.75	<0.01	0.06
	15143	658 73 ~ 657.25	10 6>	<0.01	(9.01
	15143	657 20 ~ 557.70	<0.03	<0.01	<0.01
	15150	657,70 ~ 658,20	(0.0)	(0.01	<3,91
	15151	558.23 ~ 558.70	<0.01	<0.01	<3.01
	18152	658,73 ~ 659.05	0.01	<0.01	<9.01
			<u> </u>	<u></u>	

i j	Width	Depth	T-Cu	AS-Cu	T-Co
	(1)	(•)	(1)	(1)	(1)
	1955.82	554.35 ~ 655.30	1,71	<0.01	0.05
1.1	1,45	555,30 ~ 555,70	0,49	<0.01	0.07

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Table 2-5-3 Results of Chemical Analysis of Ore Samples (3)

MJZC-12

Samole	Depth	T-Cú	AS-Cu	1-60
No.	(=)	(1)	(0)	0
KC16153	652.37 ~ 652.87	<0.01	<0.01	<0.01
KG16154	652 87 _ 653 37	<0.01	<0.01	(0 01
KC16155	653.37 _ 653.87	(0.01	(0.01	<0.01
KC16156	653 87 ~ 654.37	<0.01	<0.01	(0.01
KC16157	654 37 _ 654 87	0.09	(0.01	<0.01
KC15158	654 87 ~ 655 17	(0.01	(0.01	(0.01
XC18159	655 17 ~ 655 67	<0.01	(0.01	<0.01
KC15160	655.67 ~ 656.17	<0.01	<0.01	(0.01
KC16161	656 17 _ 656 67	<0.01	(0.01	(0.01
KC15162	656 67 _ 657 17	<0.01	<0.01	<0.01
KC16163	657.17 _ 657.67	<0.01	<0.01	(0.01
KC15164	657.67 ~ 658.37	<0.01	<0.01	0.04
KC18165	659.37 _ 658.87	<0.01	(0.01	0.03
KC16166	658 87 ~ 659 37	<0.01	(0.01	0.02
KC16167	659.37 _ 659.87	(0.01	(0.01	0.01
KC16168	659.87 ~ 660.37	<0.01	<0.01	0.01
KC16169	660 37 ~ 660 87	<0.01	(0.0)	0.01
KC16170	660 87 _ 661 46	<0.01	(0.01	0.04
KC16171	661.46 5. 651.95	1.49	(0.0)	0.11
KC16172	661.96 ~ 662.46	0.01	<0.01	0.02
KC16173	662 46 _ 662 95	0.33	(0.0)	0.02
K¢16174	662 96 _ 663 46	0.29	<0.01	0.01
KC16175	663.46 ~ 663.95	0.18	<0.03	0.01
KC16176	663.96 ~ 664.37	0.29	<0.01	0.02
KC16177	664.37 🚙 664.87	0.07	<0.01	0.01
KC16178	664 87 _ 665 37	0.12	<0.01	0.02
KC16179	665 37 🗻 655 87	0.18	(0.01	0.02
KC16180	665.87 👡 656.37	0.12	<0.01	Û. Q1
KC16181	665.37 🔔 656.87	0.14	<0 01	0.01
KC16182	666 87 _ 667 37	0.13	<0.01	0.01
KC16183	667.37 _ 687.87	0.29	<0.01	0.01
KC16184	667.87 ~ 668.37	0.03	<0.01	0.02
KC16185	668.37 ~ 668.87	0.28	(0.01	0.01
KC16186	668.87 _ 669.14	0.34	<0.01	0.02
K¢16187	669 14 🧫 669 64	1 19	(0.01	0.03
KC16188	669.64 ~ 670.14	0.54	<0.01	0.01
KC16189	670.14 🗻 670.45	0.50	<0.01	0.02
KC16190	670.45 _ 670.95	0.65	<0.01	0.03
KC15191	620.95 _ 671.45	0.66	<0.01	0 01
KC16192	671.45 ~ 671.95	0.85	(0.01	0.01
KC16193	671.95 - 672.24	0.98	(0.01	0.02
KG16194	672.24 . 672.74	2.86	(0.01	0.02
KC18195	672.74 _ 673.24	1.30	(0.0)	0.02
KC16196	673 24 ~ 673 74	2.68	<0.01	0.10
KC16197	673 74 ~ 674 14	0.45	<0.01	0.02
KC16198	674.14 🔔 674.64	0.01	(0.01	<0.01
KC16199	674.64 _ 675.14	<0.01	<0.01	<0.01

Sasple No.	Depth (s)	1-Cu (%)	AS-CU (S)	1-Co (1)
KC16200	675.14 ~ 675.64	<0.01	<0.01	<0.01
KC12201	675.64 🔔 676.14	<0.01	<0.01	<ò.01
K¢12202	676 14 ~ 676 45	<0.01	<0.01	<0.01
KC12203	675.46 ~ 676.95	<0.01	<0.01	<0.01
K¢12204	676.96 ~ 617.46	<0.01	<0.01	<0.01
KC12205	677.46 ~ 677.96	<0.01	<0.01	<0.01
KC12206	877.96 ~ 678 76	<0.01	<0.01	<0.01

Tidth	Depth	T-Cu	AS-Cu	I-Co
(4)	(m)	(%)	(1)	(6)
3.10	669.14 ~ 672.24	0.77	<0.01	0.06
1.50	672.24 ~ 673.74	2.28	<0.01	0.05

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Chapter 5 Ore Reserve Calculation

5-1 Objective

The objective of the ore reserve calculation is to assess the mineral potential of the survey area.

5-2 Method

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The calculation was carried out by ZCCM using LYNX computer of Canadian LYNX GEOSYSTEM INC.

Kriging method, Inverse Distance Squared method and manual calculation on orebody sections were studied for the ore reserve calculation. In the Kriging, borehole data points may not be enough for constructing a reliable semi-variogram, while the manual calculation overestimated the tonnage and the grade in the part of low density of borehole. Consequently Inverse Distance Squared method was adopted to the ore reserve calculation of the area.

In the Inverse Distance Squared method the grade of a block is calculated that:

$$X = \sum_{i=1}^{N} (xi/di^2) / \sum_{i=1}^{N} (1/di^2)$$

X : block grade

xi: grade composite value of neighbouring sample point (intersection grade of drill hole etc.)

di: distance between centre of block and sample point

i : neighbouring sample point

N : number of samples used for the estimation

Inverse Distance Squared method was used under the following conditions.

3D GRID MODEL DIMENSIONS: 150 x 150 x 1300 (m) SEARCH ELLIPSOID DIMENSIONS: 800 x 800 x 800 (m) CUT-OFF GRADE: 1% T-Cu ORE DENSITY: 2.67

Assay results of gold and silver for the intersections of drill holes are listed in the appendices. These results are generally low (in the order of ppb). However, there are several relatively high grade ores in some part of the Southern Area Shoot, Northern Area Shoot and in the western part of the survey area. The matter how to treat these gold and silver assays in the calculation of ore reserves are now being discussed by ZCCM.

5-3 Results

68 boreholes were found to have intersections of 1% Cu mineralization.

The orebody being mostly gentle slope lying, each 3D Grid cell created one orebody intersection as seen in the plan.

Figures and tables of various kinds on the results of the calculation are shown in appendices.

ZCCM has made a policy not to use the word ore unless an economic evaluation has been made, and concluded that the tonnage and grade of the results be expressed under two headings:

(A) Potentially Economic Mineralization: This will be summation of blocks which have a minimum true thickness of 3m and a minimum block grade of 2% T-Cu. The blocks should also be connected with each other making a minable body. This criterion was used to quantify the Northern Area Shoot and the Southern Area Shoot.

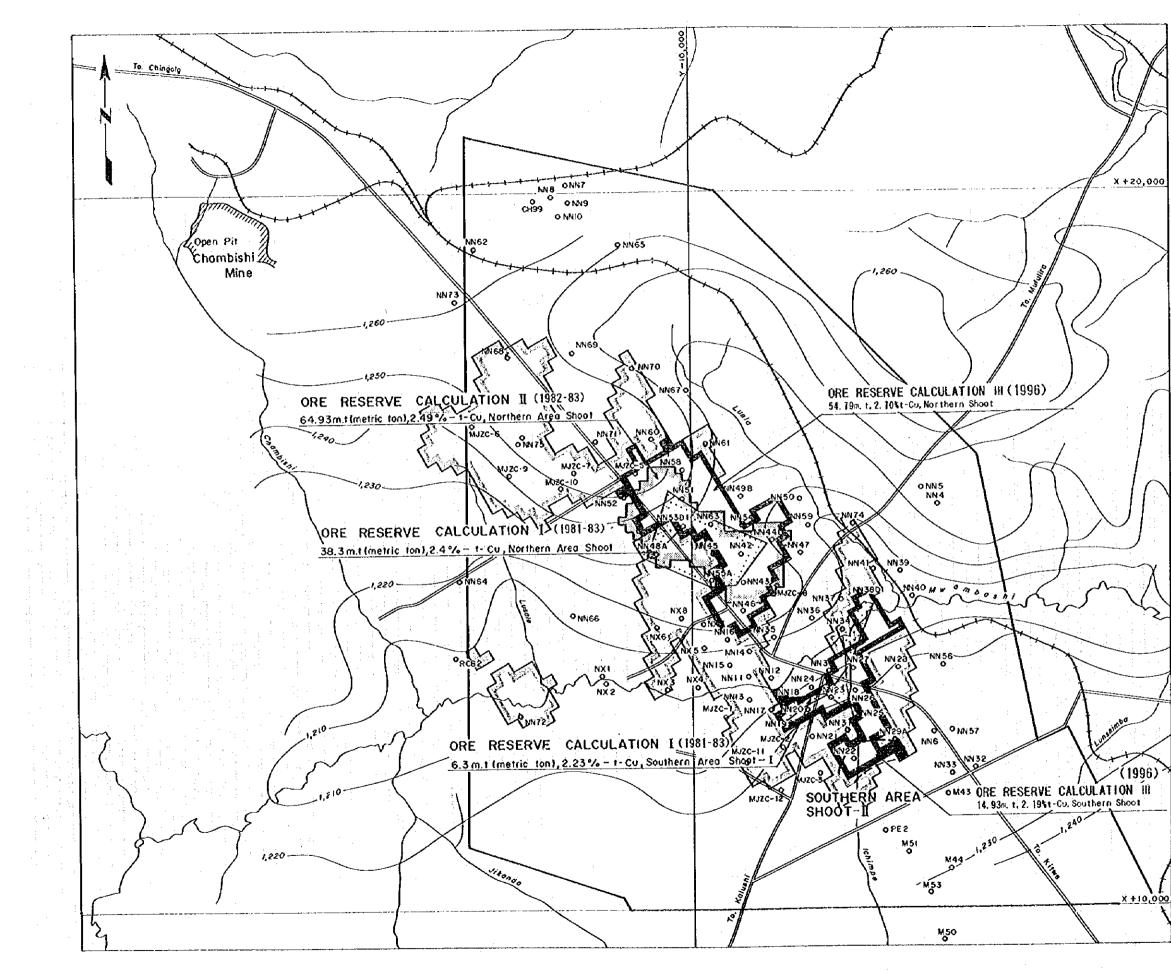
(B) Subeconomic mineralization: The grade and tonnage of the remaining blocks of the 1% Cut-Off mineralization.

The areas around NN-75, MJZC-9(NN-84) and RCB-2 were purposely left out from the Potentially Economic Mineralization as those are separated from the Northern Area Shoot and require further drilling to firm up the block grades and tonnages. However those areas are regarded as of considerable promise for location of economic mineralization.

The tonnages and grades of the survey area are as follows.

POTENTIALLY ECONOMIC MINERALIZATION;

NORTHERN AREA SHOOT: 54,793,000 tons, 2.70% T-Cu, 0.13% T-Co SOUTHERN AREA SHOOT: 14,934,000 tons, 2.19% T-Cu, 0.13% T-Co



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Dritting Holes

Topographic Elevation 1250 Contour in Metre

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Survey Areo

ORE RESERVE CALCULATION III (1996) Subeconomic Blocks, 10, 91m. t, 1, 83%t-Cu

Northern Area Shoot

	Trua Thickness	Total CuX	Total CoX
NN58	22.92	2.21	0.09
51	14.21	2.68	0.06
48-B	4.67	2.07	0.02
53-D1	4.92	2. 15	0.05
63	18.41	2.11	0.21
45	10.39	2.32	0.05
42	16.27	2. 29	0.10
44-D1	15.90	2.86	0. 18
55-A	3.02	2.04	0.04
43	12.02	2.93	0.09

Southern Area Shoot- I

	True Thickness	Total CuX	Total CoX
NN11	5.49	1.88	0.04
NN18	4, 48	2.81	0.07
20	5,06	1. 92	0.13
23	4. 75	2.62	0.27
26	4.63	1.87	0.12
27	5.12	2.31	0.28
38-D1	3.90	2.98	0.01
40	9. 78	2. 17	0.04

Southern Area Shoot-II

	True Thickness	Total CuX	Total CoX
NN22	5.61	2.37	0.13
29	9.08	1.75	0. 17

Nothwestern Area

	True Thikness (*)	Total Cu%	Total Co%
NN75	10.72	2.11	0.09
₩JZC-9	5.79	3, 12	0, 08

Fig. 2-5-1 Ore Reserve Calculation

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SUBECONOMIC MINERALIZATION (includes isolated patches of 2% Cu and 3m true thickness blocks): 107,909,000 tons, 1.83% T-Cu, 0.03% T-Co

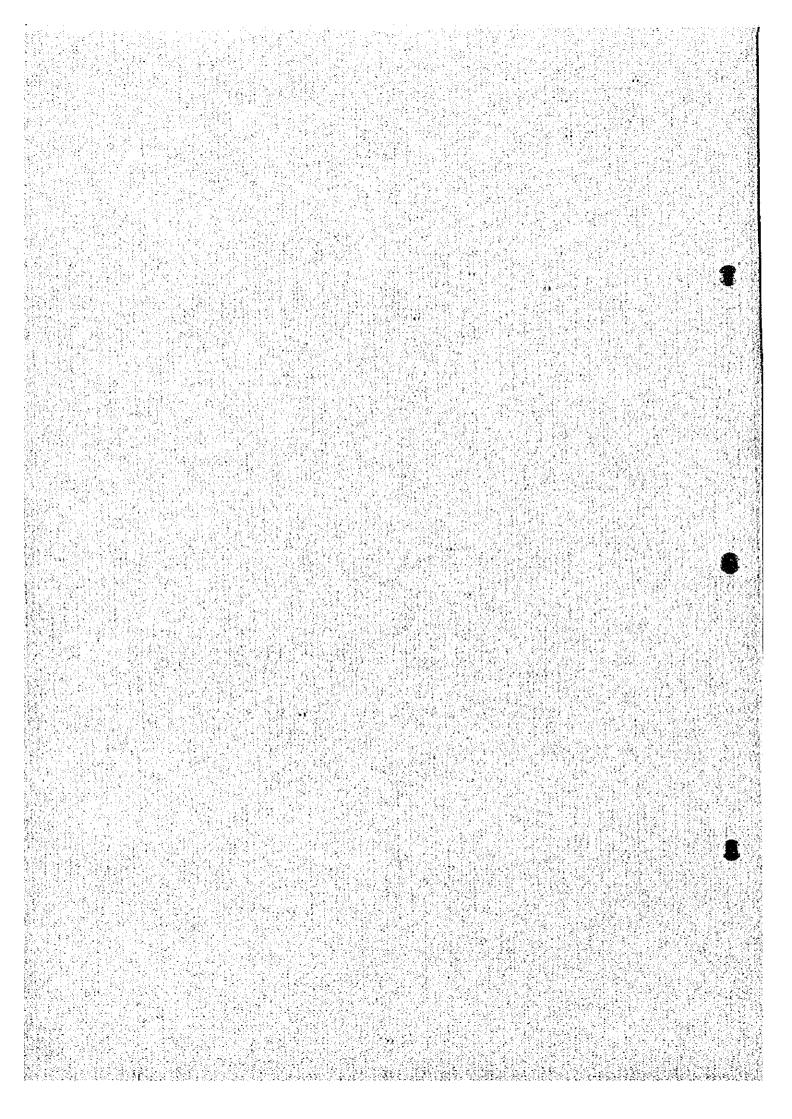
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PART III CONCLUSIONS AND RECOMMENDATIONS

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PART III Conclusions and Recommendations

CHAPTER 1 Conclusions

Drilling was carried out during the third phase of the Chambishi Southeast survey. All four holes drilled (MJZC-9, -10, -11, -12) this year accomplished their objective by penetrating the ore horizon. Of these four, three reached the basement. The results of these work clarified considerably the geology and mineralization of the western and southern parts of the survey area and the following conclusions were attained.

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MJZC-9 drilled in the western part of the area confirmed 1. grade the existence of high-grade ores (i. width 5.90m, T-Cu T-Co 0.08%; ii. width 2.58m, grade T-Cu 2.29%. T-Co 3.12%. These ores are considered to be continuous to the <0.01%). shoot confirmed to the north of this hole (NN-75). is Thus it now clear that an ore shoot of considerable scale exists in this area. It is inferred that this ore shoot is emplaced over a basement depression which is elongated in the NE-SW direction and it is deemed possible that this shoot is developed further southward or westward.

2. Relatively low-grade copper ores rich in pyrrhotite were confirmed by MJZC-10 drilled on the eastern side of MJZC-9 and $MJZC-11 \cdot -12$ drilled in the southern part of the area. In these mineralized zones, the rich ore probably deposited during a relatively short period of time before or after the start of deposition of the "Ore Shale", because Fe/Cu ratio tends to increase upward.

3. Small scale bornite-chalcopyrite mineralization was confirmed in the basal part of the "Ore Shale" in MJZC-12. It is inferred that this mineralized zone lies over a basement-rise which extends southward. Palaeo-basement highs probably existed in parts of this rise at the time of ore deposition. Therefore, palaeo-basement highs occurred in relatively shallow parts and there are possibilities of ore shoots occurring on their limbs.

4. A basement depression extending in the NB-SW direction is considered to exist between MJZC-11 and -12. This should be noted as MJZC-2 with relatively high grade ores is located on

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the northeastern extension of this depression.

5. Ore reserve estimation was carried out to asses the mineral potential of the survey area with the following results.

POTENTIALLY ECONOMIC MINERALIZATION;

NORTHERN AREA SHOOT: 54,793,000 tons, 2.70% T-Cu, 0.13% T-Co SOUTHERN AREA SHOOT: 14,934,000 tons, 2.19% T-Cu, 0.13% T-Co SUBECONOMIC MINERALIZATION (includes isolated patches of 2% Cu and 3m true thickness blocks):

107,909,000 tons, 1.83% T-Cu, 0.03% T-Co

CHAPTER 2 Recommendations for Future Exploration

Significant amounts of ore were confirmed in this survey area by drilling during this year. The ore deposits of this area, however, occur in relatively deep zones, the major deposits probably occurs at 550 to 1,050m below the surface. Therefore, in order to develop this deposit, it is necessary to further increase the ore reserves. The western and southern parts of the survey area have not been explored and the potential is considered to be promising.

now clear, from the results of the present survey, is Tt that a deposit which was hitherto unknown occurs in the western part of the area. Also borehole RCB-2 which previously confirmed ores is located far south of MJZC-9 which also confirmed ores. From the above it is strongly recommended that efforts be concentrated as follows to confirming new ore reserves and to exploring the vicinity. First drill at sites where the depth of the ore deposits can be estimated at shallow depths, namely near the two boreholes which encountered ores (MJZC-9, NN-75). drill at sites where the depth of the ore is considered then to become deeper, namely south and west of MJZC-9.

The possibility of ore shoots still remain in the southern part of the area and thus it is recommended that drilling be carried out in the area to the south of MJZC-12.

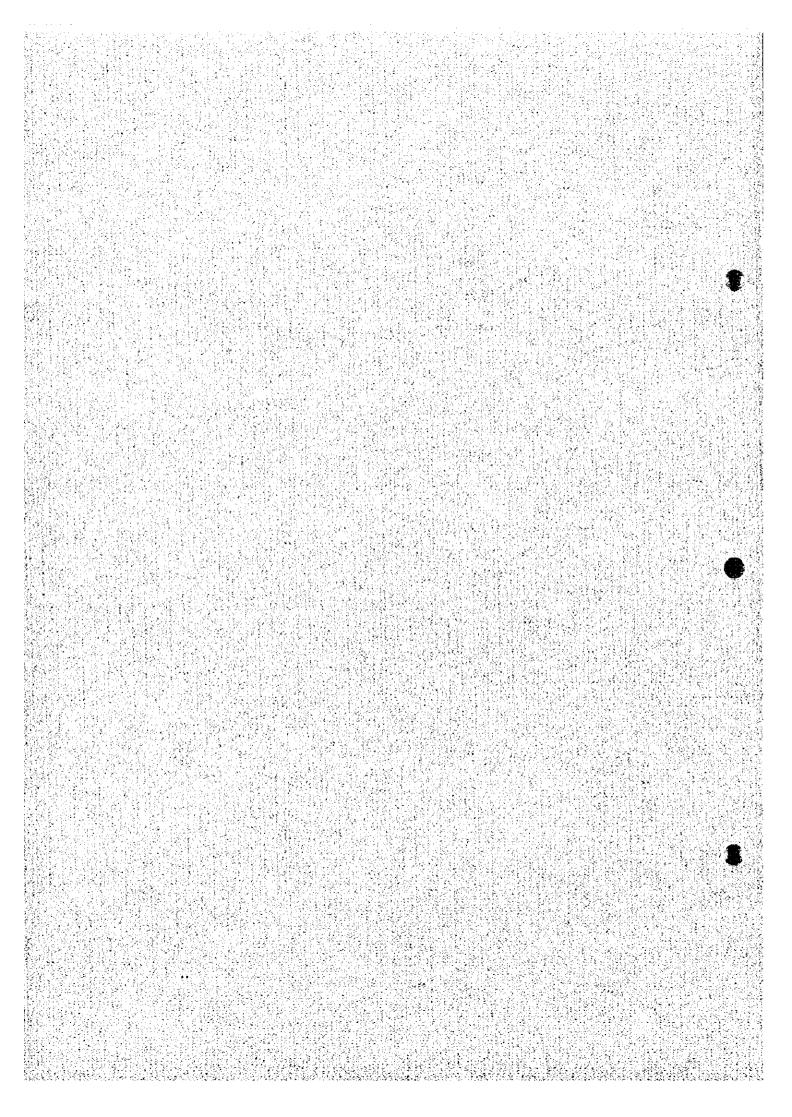
Also in order to accurately determine the ore reserves of the Northern Area Shoot, the main deposit, drilling should be carried out near the peripheries of the deposit.

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