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SUMMARY REPORT

NEAR-SURFACE COPPER-MOLYBDENUM SULFIDE MINERALIZATION AT COPPER CREEK

including

Mineral Reserve Calculations

A.J. Perry R.B. Blakestad Perry, Knox, Kaufman, Inc.

March, 1975

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PURPOSE

SW 70 Participants have been engaged in exploration at Copper Creek since 1970.

The purpose of this report is four fold: (1) to summarize briefly the results of recent shallow rotary-percussion drilling at that location, (2) to calculate a reserve of near surface mineralization after combining recent results with data from previous area exploration, (3) to comment on the possible economics of exploitation of the reserve and (4) to speculate briefly on possible remaining deeper copper potential -- making suggestions as to "farmout possibilities".

SUMMARY AND CONCLUSIONS

Recent rotary-percussion drilling conducted at Copper Creek verified the presence of two near surface bodies of sulfide copper and molybdenum. Our rough calculations show a reserve of +20,000,000 tons of material at a grade of 0.42% Cu eq. The waste to ore ratio is .92:1.

Assuming a 5000 tpd open pit operation, a conventional sulfide concentrator and a 65ϕ NSR copper price, \$87.4 million would be generated over a 12 year operational life.

An increase of 20¢ in copper price would raise total revenues by 23.5% but the average ROI would be elevated to only 11.6%. Other indicators are also unfavorable.

Additional exploration might add reserves to those calculated but it is doubtful that the increase would equal 20% of the copper now outlined. An increase in grade should not be expected.

It is concluded that supergene copper reserves at Copper Creek are unlikely to sustain an economic operation in the foreseeable future.

Additional deeper tests for primary copper-molybdenum sulfides appear warranted.

RECOMMENDATION

It is recommended that ERC maintain their current mineral interests at Copper Creek (including the Gillings and Trust 60 optioned lands) for a period of 60 days. During this time an attempt should be made to interest the several parties who have inquired as to the availability of these lands for exploration, and others, in undertaking further exploration. The best arrangement would appear to be one where ERC-PKK would have an opportunity to "back in" after expenditures to date had been equalled.

LOCATION

ERC-PKK Copper Creek properties, held under terms of the SW 70 Agreement, are situated in Secs. 23, 2^4 , 25 and 26, $T12\frac{1}{2}N-R3W$, Yavapai County, Arizona. Copper Creek (Little Copper Creek) is located in an area of rugged low hills just south of Hwy 89 about 12 road miles SW of Prescott, and $3\frac{1}{2}-4$ miles SE of Phelps Dodge's Copper Basin copper deposit. All Project lands are within the Prescott National Forest. See the following portion of the Kirkland Quad (Figure 4) for location.

BACKGROUND

The presence of multiple intrusives of Laramide age at Copper Creek with attendant breccias, pervasive alteration and disseminated copper-molybdenum mineralization has long been recognized. However, the absence of strong "live limonite" indicating substantial enrichment was responsible for the lack of meaningful exploration until Phelps Dodge drilled several core holes in about 1963-64. There followed a period of occasional occupancy and geologic work by several groups but no additional drilling was done until 1970-71 when two groups, SW 70 participants (then ERC, Day Mines and PKK) and Norandex, Inc. - Sierra Mineral Management (SMM) simultaneously occupied lands covering the indicated pervasive mineralization.

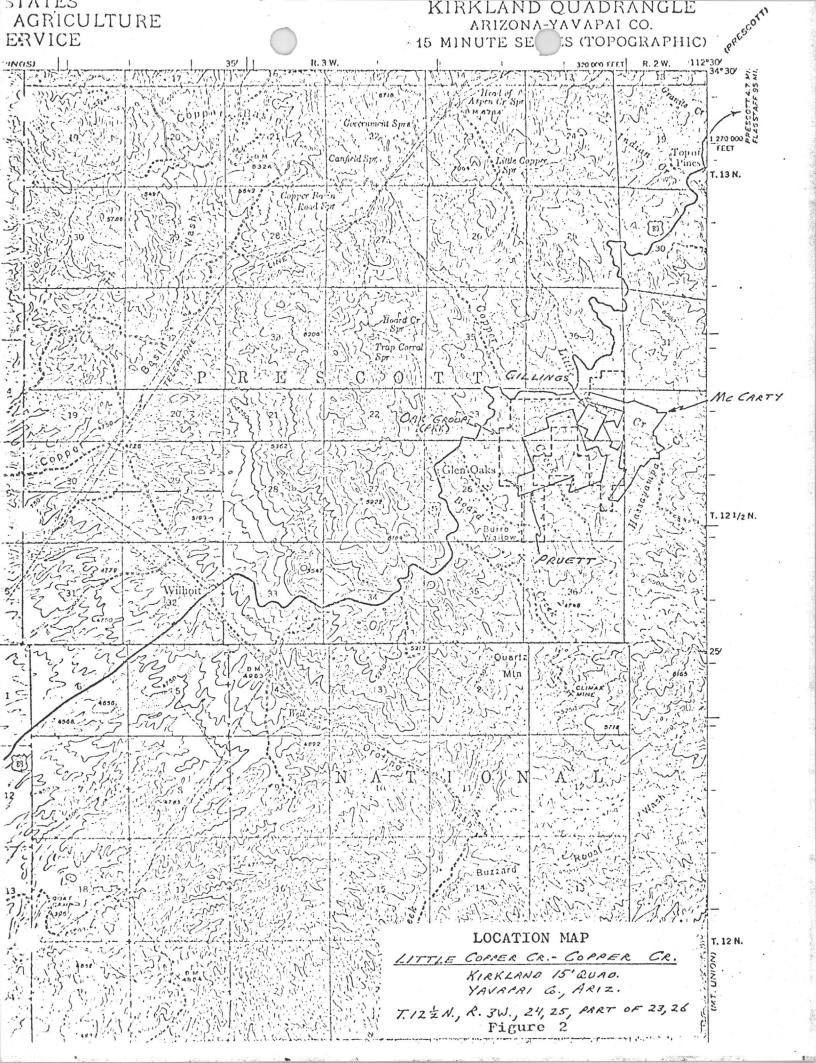
The Norandex - SMM group subsequently abandoned their efforts. SW 70 optioned their lands to Essex International in the fall of 1973. Essex drilled 9 shallow holes in search of supergene ores and promptly terminated the agreement. In March, 1974, SW 70 optioned most of the land formerly held by Norandex (Trust 60 Properties).

In August, 1974, J.N. Swinderman of ERC proposed additional exploration at Copper Creek, for both supergene ores and for the deeper sulfide potential. ERC management approved the shallow drilling project. Day Mines, a 50% financial participant in SW 70, withdrew from the Project at that time.

Exploration for additional chalcocite ores, the principal subject of this report, commenced December, 1974.

RECENT SHALLOW ROTARY PERCUSSION DRILLING

The recent rotary percussion drilling was undertaken at Copper Creek to further test areas designated by Swinderman as potential enriched areas amenable to open pit mining (North and South areas of Figure 1). The CC74 series holes of that Figure are the result of that drilling.



A total of 2929 feet was drilled (including minor coring) in 12 holes. Abbreviated descriptive logs, simple geologic strip logs, a tabulation of assays, and a description of sampling procedures (all for CC74 series drilling) are appended.

The recent drilling has established that a small reserve of low grade supergene copper mineralization with some by-product molybdenum, all amenable to conventional flotation, is present near surface in two areas on the properties currently held by ERC-PKK.

Discussions as to the character of the mineralization, alteration, limitations of available data, and an "ore" reserve calculation as well as some comments on potential economics follow.

CHARACTER OF THE MINERALIZATION AND ALTERATION

Practically all pre-mineral rocks intersected by the rotary-percussion drilling are quartz-rich (Qmp-qlp-etc.) and low in total sulfides -- generally 1-3% or less.

Pyrite is the predominate sulfide. There are, in order of decreasing abundance, lesser amounts of chalcocite, chalcopyrite, molybdenite and covellite. Bornite is occasionally present.

The principal copper minerals are chalcocite and chalcopyrite. The former occurs as coatings on pyrite and chalcopyrite, both of which occur primarily as discrete disseminations -- with only minor amounts being found as veinlets or fracture coatings. MoS₂ also occurs primarily as discrete flakes either in silicified country rock or along quartz vein selveges. There is no direct relationship between the better MoS₂ sections and the better copper intervals (see assay tabulation, appendix).

The supergene copper mineralization appears to have little relationship to any particular pre-ore Laramide rocks tested by this drilling -- the degree of chalcocite development being largely a function of the amount of chalcopyrite originally present -- and that in turn having been determined by proximity to available copper bearing solutions (centers of mineralization, more well broken areas, etc.).

Reference to the strip logs, cross sections and assay tabulations attached will show that the downward sequence of mineralization is normally leached capping, chalcocite zone and primary mineralization, in holes drilled from topographic prominences. Holes collared in gulch bottoms generally show sulfides almost immediately -- with only minor oxides being present. Oxide grades in the better mineralized areas are generally .01-.03% Cu grading rapidly upward as chalcocite-coated sulfides are encountered.

Chalcocite, when present, is generally low-grade -- with individual $2\frac{1}{2}$ ' or 5' assays rarely exceeding 0.50% Cu. The highest copper assay recorded was in CC74-3 (70-75' = 1.09% Cu) - primarily a chalcocite section.

Sulfide grades of 0.10-0.15% Cu are often found below good chalcocite. Lesser protore grades are common outside the principal areas of mineralization.

Silver commonly reported to assay in amounts between .02-.08 oz/T, and occasionally exceeded.10 oz/T. The specific silver mineralogy is not known but we presume the silver is tied to some of the copper minerals. However, there is no direct relationship between the higher copper intervals and high silver. Essentially no gold has been reported from the few assays collected from previous drilling at Copper Creek.

Qualitative spectrographic analyses of three samples from CC74 series cuttings showed no other elements reporting in amounts suggesting their possible economic interest.

Chalcocite Mineralization

Karvinen of Norandex, Briscoe of Sierra Mineral Management, J. Swinderman of ERC, and, in recent months, this author believed that evidence from "live limonite" capping and drill hole analyses available suggested that enriched ores (a chalcocite blanket) of economic interest might be available on ERC-PKK properties at Copper Creek. Results of recent drilling combined with previous exploration results has made possible the reserve calculation of this report.

"Live limonite" is present in discontinuous patches in the area of our recent drilling in Secs. 2^4 and 2^5 (North and South areas of Figure 1, attached) and locally elsewhere on the property. The favorable limonite colors are intermixed with those suggesting 1-5% pyrite. Some minor pitch after chalcopyrite is also occasionally observed. Turquoise is present here and there in thin $(<\frac{1}{2}$ " thick) stringers and blebs. Ferrimolybdite is suggested by occasional pale yellow colorations, especially in well-silicified and veined areas, where limonite after pyrite does not obscure the delicate yellow ochre.

Alteration

Reference to the abbreviated descriptive logs of CC74 series holes (Appendix) will show propylitization and silicification as being the dominant alteration types encountered. Some secondary biotite was identified. Minor K-spar development was logged. Sericite and argillite is probably generally more well-developed than recognized in our hand lens/binocular

microscope examination of wet CC74 series cuttings.

Silicification is particularly well-developed in the North Area -- perhaps to the extent of making a substantial difference in any anticipated excavation.

The obvious overlapping ("telescoping") of alteration types was early recognized by Sayers of PKK and supported by the limited petrographic work accomplished.

RESERVE SUMMARY

Reserves of copper/molybdenum-bearing material available in the two areas outlined as North and South areas on Figures 2 and 3, using a 0.25% Cu eq. exterior cutoff (0.20% Cu eq. interior cutoff) are as follows:

North Area - 12,993,280 tons - 0.428 Cu eq.

South Area - 8,371,520 tons - 0.410 Cu eq.

Grand Total- 21,364,800 tons - 0.42 Cu eq.

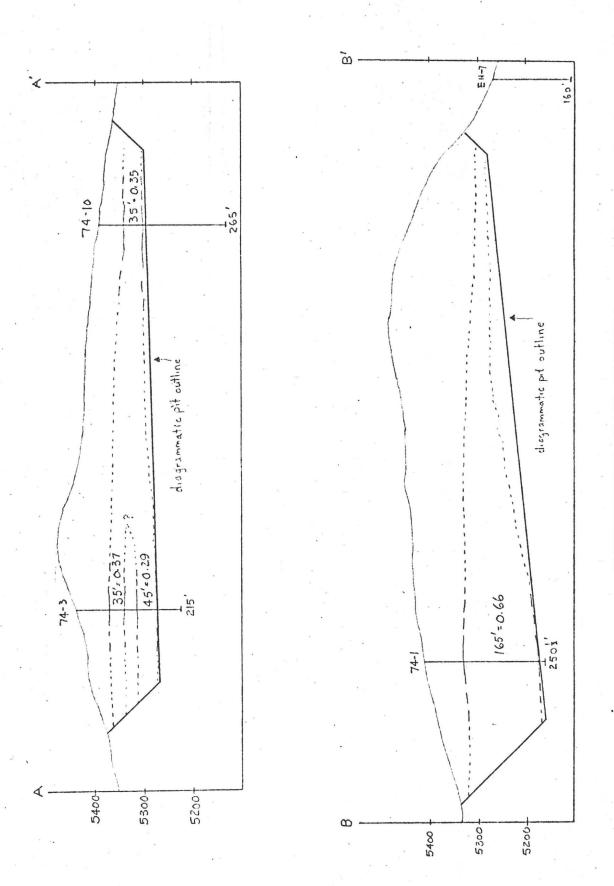
Total waste measured, both areas, was 19,587,424 tons, resulting in a waste (all categories) to ore ratio of 0.92:1.

You are referred to the appendix for the detail and further summary of the reserve and waste calculations.

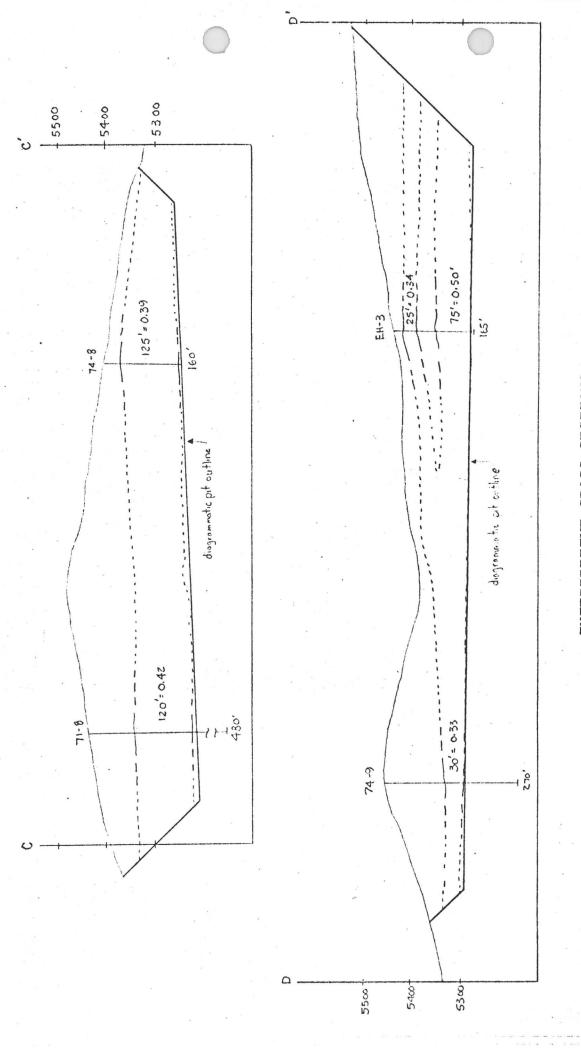
OREBODY CONFIGURATIONS

Figure 2, used as an overlay to Figure 1, will show that the North Area "orebody" has dimensions of approximately 1800' X 1100'. Ore thicknesses vary from 10 to 185 feet. Sections A, B, C and D show that the ore is generally flat lying and irregularly tabular, with a slight westerly dip. The ore (using a 0.25% Cu eq. cutoff) tends to be thicker on the west side of the body. There is a tendency for the better zone to drape downward with decreased surface elevations. Chalcocite was measured in the top of DDH CC-4 located just outside the west edge of North Area oreblock (but inside the proposed pit), illustrating that chalcocite can extend beneath valley bottoms, though this may be an exception.

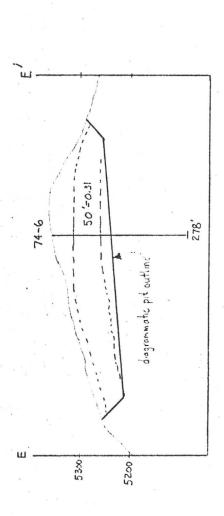
The South Area (use overlay, Figure 3) is an irregular ellipse measuring approximately 1600' X 700' in plan. Ore thicknesses vary from 35 to 160 feet. Sections E, F, G and H illustrate interpreted ore configurations. A particularly good thickness of chalcocite is measured in HA-6, which was collared on a topographic prominence.

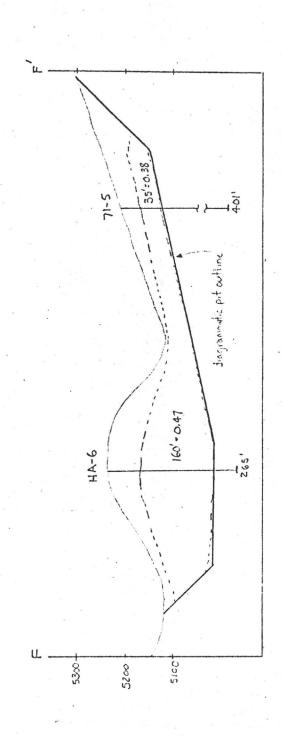


INTERPRETIVE CROSS SECTIONS
NORTH AREA
Figure 5

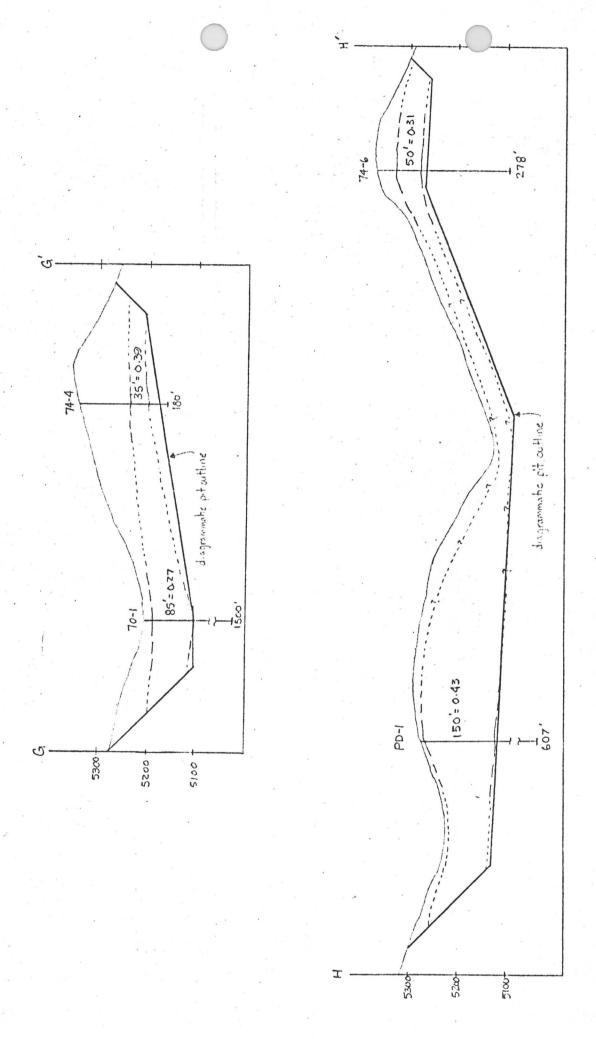


INTERPRETIVE CROSS SECTIONS
NORTH AREA
Figure 6





INTERPRETIVE CROSS SECTIONS
SOUTH AREA
Figure 7



INTERPRETIVE CROSS SECTIONS
SOUTH AREA
Figure 8

RESERVE CALCULATION PROCEDURES

Cut-off grades for ore block limits and interior ore of 0.25% Cu eq. and 0.20% Cu eq., respectively, were established by accepted methods -- considering a metals market price of $68\phi/\#\text{Cu}$. Modifications of costs (milling overhead and other costs upward) after commencement of the calculation resulted in increasing the copper price to fit the case for report purposes. The final NSR price used for economic considerations was $65\phi/\#\text{Cu}$.

Reference to Figures 2 and 3 (attached) will show the distribution of drill holes within "orebody" limits in both the North and South areas to be irregular. A cross section calculation scheme was studied and rejected in favor of a polygonal system using lines connecting midpoints between adjacent holes, modified for terrain and geology. Several cross sections, Figures 5 thru 8 are provided to show suggested ore continuity.

In general, orebody perimeter limits were extended 150 feet or less beyond the outermost hole considered for pit incorporation. Exceptions were made, as in the case of CC-3 in the North Area, where topography and an unusual ore thickness dictated an outward extension. Hole PD-2 (South Area) was rejected from ore calculation.

Polygon areas were each planimetered 4 times and averaged. The thickness of ore and overburden for each polygon was extended over the block from the controlling hole for calculation purposes. A 12.5 Cu ft/ton factor was used in the tonnage determinations.

Pit walls were sloped at -45° from ore limits. By use of cross sections, the surface orebody limit was projected vertically downward to the elevation of the bottom of ore (with some modification for chalcocite zone draping) and a +45° line extended outward from that point to surface. Volumes of pit layback material were calculated using an arithmetic average of cross sectional areas and a planimetered surface area.

Tables of tonnage and overburden calculations are appended.

EVALUATION-DATA LIMITATION

Considering the purpose/sensitivity of this evaluation there appears to be one substantial limitation to data used for reserve calculations. That is the combining of analyses from several sources -- each of a different reliability. Four Phelps Dodge holes (PD series) and those of Norandex (HA, 70 and 71 series holes) analyses were collected by Swinderman

of ERC from Sierra Minerals graphic assay representations. Essex (EH series) holes were rotary holes with analyses by Iron King. Check assays of select rejected cuttings indicated these analyses were reliable. Molybdenite analyses for many of the substantial copper intercepts in these holes are not available.

Portions of CC74 holes (those of ERC's recent drilling) were analyzed for total copper by Hawley and Hawley; other intervals by ERC. There was generally good agreement. ERC analyzed for copper oxide and molybdenum as well. Checks for moly by Rocky Mtn. Geochem-Tucson indicated ERC's reports of molybdenum were on an average 20% high. All ERC Mo analyses have been subsequently reduced before being incorporated on Figure 1 and in reserve calculations. All assays from CC74 series drilling are tabulated in the Appendix, as reported, without any modification.

METALLURGICAL CONSIDERATIONS

Sulfide Flotation

Hazen Research, Inc., of Golden is conducting preliminary flotation tests on a composite of cuttings from some of the better-grade recent CC74 series holes intersections. Only very preliminary results are now available. They are somewhat encouraging.

Ralph Light, Hazen VP, reports by phone that using a 0.41% Cu head sample and a grind, allowing 70-80% to pass -200 mesh, about 80% of the total copper available (includes both sulfide and the minor oxide present) is recovered in an unwashed rougher bulk sulfide concentrate. About 90% of the total sulfide copper is being recovered. The bulk concentrate assays about 5% Cu and 0.4% Mo.

Some preliminary Hazen remarks suggested anticipated difficulty with copper sulfide - molybdenite separation.

Leaching Possibilities

The Hassayampa Project Report of Sierra emphasizes their approach to the near-surface Copper Creek reserves as a "leachable" blanket. Support for this thesis came from consultants J. Still and Dave Lowell.

Using the better sections of CC7 4 series holes for which there are copper oxide assays (ERC Lab), the arithmetic average of CuOx to total copper shows that an average of $13\frac{1}{2}\%$ of the total copper measured reports as oxide copper. This varies from 3% (CC74-1; 168-248) to 35% in CC74-11; 20-30).

Hawley and Hawley, Tucson assayers, upon instructions to determine the amount of readily leachable copper - not necessarily just copper in oxide form -- measured 59% and 66% of the contained copper as leachable -- but leachable only in a very controlled environment: in the case of the 66% soluble copper, only after a 5% H₂So₄ solution was brought to a boil for 2 minutes (see attached certificate of analysis - Appendix)

In summary, less than 15% of the total copper available in the near-surface mineralized sections tested at Copper Creek reports as oxide copper which would be readily available for leaching. This amount is insufficient to consider recovering either on: (1) a strictly leach ore basis or (2) leaching in combination with sulfide flotation.

Only conventional flotation of a copper and a molybdenum sulfide concentrate are considered in this summary.

MINING CONSIDERATIONS

Substantial portions of the waste and ore from both North and South area model pits would be removed from above the current topographic base levels if mining were undertaken.

Lands in addition to these currently held would be necessary for some waste disposal and for plant site and tailings impoundment.

Rocks at Copper Creek are of a normal porphyry type, locally well fractured, but hard. Drilling and blasting will be necessary from the onset of stripping. Considering the small working areas and the normal demands of flexibility of a small operation, a loader and truck operation should be considered.

The perimeter of the proposed pit of the North Area orebody would be within 1200 feet of Highway 89; making ecological considerations important.

ECONOMIC ANALYSIS

Following is a brief economic run-thru using as a model W.C. Coles' analyses of October, 1974*.

Using the reserves of this report, assuming open pit mining, a 5000 tpd conventional flotation plant and a 12 year mine life, a total revenue of \$87,400,000 would be generated at 65¢ NSR value copper. The average annual ROI provided would be 5.0%.

Assuming 85¢ NSR value copper, revenue increases to \$114,200,000; but we judge other indicators fall short of ERC's minimum goal.

Reference to the tabulated drill hole analyses in the appendix will show that only thin intercepts of 0.50% Cu eq could be developed for calculation at that grade.

It would appear to PKK that the supergene reserves developed at Copper Creek are substantially less in terms of tons and grade than those required for an economic operation in the forseeable future.

It is estimated that additional exploration might increase the reserve tonnage somewhat, probably adding < 20%. An increase in the grade above that currently measured should not be expected.

^{*}Letter from W.C. Cole to Wm. M. Calhoun, Day Mines, Inc.

COPPER CREEK

Basic Assumptions:

Reserves: 20,000,000 tons @ 0.42 Cu equivalent.

Waste:Ore

.92:1

Mine Life:

12 years

Mill Operates:

325 days/yr.

Flotation Concentrator:

5000 tpd

NSR Revenue:

US Producer Price less 15¢

Capital Costs:

Roads	67,500.
Water Supply	337,500.
Mining Equipment	2,025,000.
Office	275,000.
Mill	12,500,000.
Tailings System	675,000.
Mine Shop	337,500.
Engr. Fees	540,000.
Dev. Drilling	540,000.
Pre-Production Mining	540,000.
Working Capital	810,000.
Sub-Total	\$18,647,500.
+10%	1,864,750.
Total	\$20,512,250.

Note: Assume no replacement of capital equipment and no salvage.

Operating Costs

Mining 35¢/ton of rock @.92:1 = .67 ton
Milling 1.35

G&A @ 30% .41

Royalty (2% NSR assuming 40% ore from Trust 60) .05

Total \$2.48

X = 20,000,000 tons = \$149,600,000

Revenue

Total Profit

Payback Time =

NSR FOB Concentrator = \$4.37/\$tonX 20,000,000 tons = \$87,400,000

Summary 12 year mine life @ 65¢ NSR Cu, 0.42% Cu eq. head grade...

.60

5.0%

Profit + Investment

Ave. Annual ROI

Revenue	87,400,000.
Capital costs	20,512,000.
Operating Costs	49,600,000.
Pre-Tax Profit	\$17,288,000.
Depletion (50%)	8,644,000.
Taxable Income	8,644,000.
State and Co. Taxes @ 17%	1,469,480.
Net before Fed. Taxes	7,174,520.
Fed. Tax @ 48%	3,443,770.
Post tax profit	3,730,750
+Deprec. +Depl.	29,156,000
Cash Flow	32,886,750
displace read each in winning the large structure opposition with restaurance and read of the contraction of	***************************************

\$12,374,750

7.5 years

Summary 12 year mine life @ 85¢ NSR Cu - 0.42% Cu head grade

Revenue	\$114,200,000
Cap. Costs	20,512,000
*Operating costs	50,000,000
•	
Pre Tax Profit	\$43,688,000
Depletion (15%)	17,130,000
Taxable Income	26,558,000
State & Local Taxes @ 17%	4,514,860
Net before Fed. Taxes	22,043,140
Fed. Taxes @ 48%	10,580,707
•	
Post Tax Profit	11,462,433
+Deprec. +Dep1.	37,642,000
Cash Flow	49,104,433

Total Profit	28,592,433
Payback Time	5.0 years
Profit * Investment	1.40 years
Ave. Annual ROI	11.6%

^{*0.42 = 8.4#} X .8 = 6.72# rec. X .85 = \$5.71 NSR value raw ore. Change from 80 % mkt. price calc. because of mod. of Trust 60 Royalty (.67 + 1.35 + .41 + .07 = \$2.50)

DEEPER SULFIDE POTENTIAL

A total of 52 holes are known to have been drilled on ground currently under ERC-PKK control at Copper Creek. Only three of those holes have exceeded 1000 feet in total depth: PD-6 (1235'), DDH70-1 (1500') and CC-4 (1611').

Reference to attached Figure 9 will show that the 3 deeper holes are so spaced as to form an irregular fence with a NE-SW orientation with holes spaced 1500 and 1800 feet apart.

The geology of holes PD-6 and DDH70-1 are unfortunately not known. SW70 Project hole CC4 bottomed at an elevation of 3714, penetrating sparsely mineralized Pre-Cambrian rocks at about elevation 3798. DDH70-1, SW of CC-4, bottomed at elevation 3700, presumably with no Pre-Cambrian intersected, but with no increase in copper. PD-6 penetrated only to elevation 4214, with some increase in copper (rising from 0.08% Cu ±0.14% Cu near the bottom) again with no apparent Pre-Cambrian intersection. Sierra reports that there was also an increase in alteration with depth in PD-6.

Reference to Sayers' transparent alteration outline overlays and the locations of the three deeper holes at Copper Creek will clearly show that only the central portion of the south half of the altered zone has been tested by the +1000' deep holes.

It seems reasonable to assume that a deeper penetration than the 1236' of PD-6 is warranted at that location -- in view of the mineralization - alteration increase -- particularly in light of Swinderman's hypothesis of a SW plunge toward a buried intrusive center. However, other hypotheses developed could also have merit.

PROPERTY CONSIDERATIONS

ERC-PKK currently control the following properties at Copper Creek:

- (1) Oak Claims Project located 58 unpatented lodes and fractions.
- (2) Gillings Property Two patented claims, leased with the option to purchase, covered by Agreement dated July 14, 1971. Total price for purchase of the property is \$65,000 (option period expires 1976).
- (3) Trust 60 Property Lease Agreement and Option to

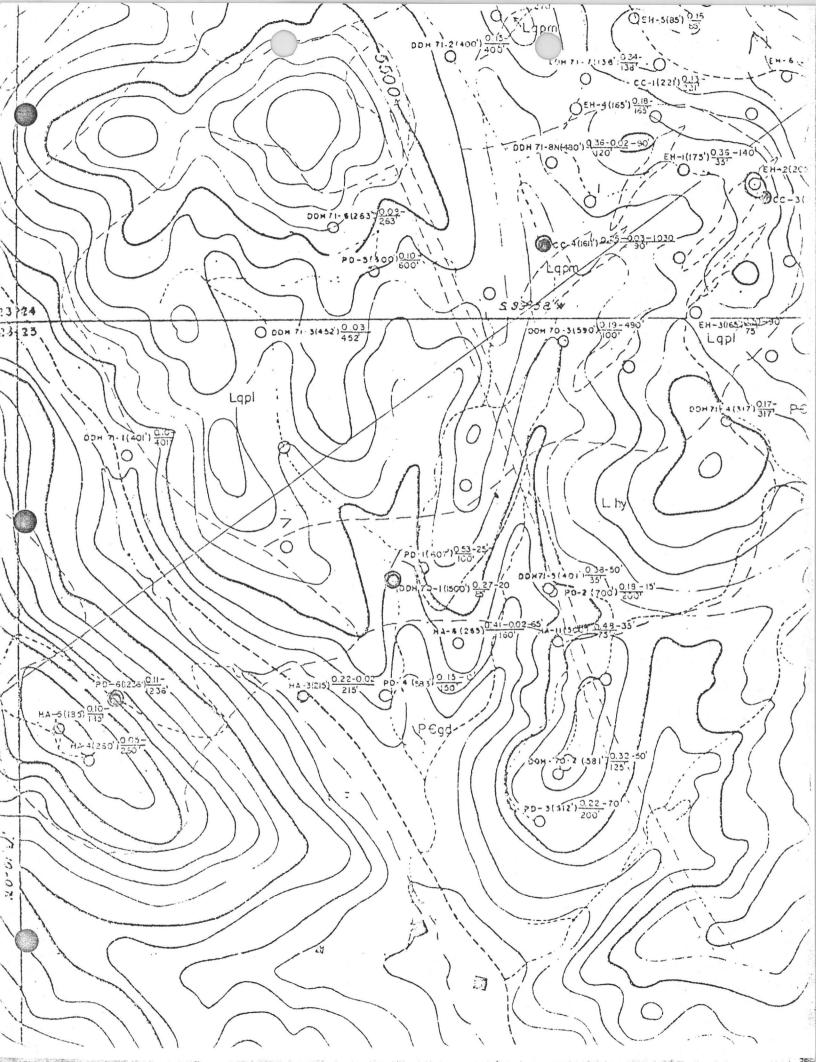
 Purchase dated March 7, 1974, covers 14 patented,

 8 unpatented lodes and 2 unpatented millsite
 claims. Total purchase price is \$5 million -until the property is purchased, a 3% NSR
 payment is due on any production.

Reference to the outlines of the supergene mineralized zones shown in Figures 2 and 3 and to a 500 scale claims map (Company files) will show that about 90% of the North Area mineral reserve is on ground optioned from Gillings; 100% of South Area mineralization is on Trust 60 lands.

Studies of the pervasive alteration indicating areas of copper potential were made by Sayers (PKK) in 1970. This study indicated that 58.3% of the favorable ground was then held by Norandex (Trust 60 property) and 41.7% by ERC-PKK (including Gillings). We now control essentially the entire altered area.

Current monthly payment obligations are: \$1650/mo. to Trust 60 and \$550/mo. to Gillings.



APPENDIX

NO.	
1	Drill Hole Assay Tabulation - CC74 Series Rotary- Percussion Holes.
2	Abbreviated Descriptive Logs - CC74 Series Rotary-Percussion Holes.
3	Graphic Logs - CC74 Series Rotary-Percussion Holes.
4	Qualitative Spectrographic Analyses - CC74 Series Cuttings.
4	Oxide (Leachability) Tests - Certificate of Analysis.
·5	Ore and Waste Calculations - North Area.
5	Ore and Waste Calculations - South Area.
5	Pit Layback Waste Calculations and Calculations of Waste to Ore Ratios.

12/74-1/75 Rotary Drilling TD - 250.5'

	Hole No.				*		
	Sample Interval	15-20	30-35	45-50	59-09	25 80 85 85 90 90 95 100	100-105 105-110 110-115 115-120
Hawley & Hawley Assay	n &	0.01	0.01	0.01	0.01	44.60	0.69 0.68 0.84 0.77
	ಬ್ ಬ				000	000000000000000000000000000000000000000	270070
A S	Oxide Cu %				000	00000000000000000000000000000000000000	00000
ERC	110				0.00.00.00.00.00.00.00.00.00.00.00.00.0	000000000000000000000000000000000000000	00000
	Ag oz/T				0.12	000000 000000 000000 000000	00000
	*				65-67.5° 67.5-70° 70-72.5°	2.5-7	

& Hawley	say
Hawley	AS

Hole No.

	*,	Ag OZ/T	0.088	0.00	000	.07	70.	0.0000	5550	0.
say	ERCASSAY	01/2	0.053	000	000	.02	10.	000000000000000000000000000000000000000	80.00	TO.
	AB	Oxide Cu %	0.024	20.0	10.0	.02	000	0.0012	484	10.
		n %	0.0000	822	627	.62	かがす	0000 0000 0000 0000 0000 0000	2222	20.
	vley									
	Hawley & Hawley Assay	Ca Ca	00.02					00000 W 4 W W W W 4 W W W W W W W W W W W W W W W W W W	からいっ	
	Hawle									
	•	Sample Interval	125-130 130-135 135-136	- シックィ - シック - ユー - ユー - ハー - シックト	60-16 65-17	70-17	80-18 85-19 90-19	200-205 205-210 210-215 220-220	30-23	ナンーのか

12/74-1/75 Rotary Drilling TD - 250.5'

Hole No.

				(85-87.5")	(95-97.5")	
	Ag OZ/T		00000000000000000000000000000000000000	5555	555	0.073
ERC Assay	110		000000000000000000000000000000000000000	2000	000	0.022
A A	Oxide		0.013	4000		0.004
	5, ₁₂		00000000000000000000000000000000000000	JW48	224	0.225
wley						
Hawley & Hawley Assay	Ca	000000000000000000000000000000000000000	44444	J.W. 43.C	3	0.24
	Sample Interval	4998919 6040501 1111 11111 1111 11111 11111 11111 11111 11111 11111 11111 11111 11111	2 2 2 2 2 2 2 2 2 2 2 2 2 2 2 2 2 2 2	000) 	100-105

12/74-1/75 Rotary Drilling TD - 250.5'

		, וא מני אנו)	((140-142.5")							
	Ag oz/T	640.0	0.079	0.073	67000	6470*0		,				
ERCASSAY	o E	10.	0000	10.	10°	0.013						
As	Oxide Cu %	0 0	0.00	00	00	0.008						
	7. C	0.175	0.195	0.220	0.220	0.185				•		
Hawley & Hawley Assay	Cu %	0.19	0.20	0.24 0.18	0.23	0.21	0.13	0.18	20.0	0.13	90*0	0.10
	Sample <u>Interval</u>	110-115	115-120	125-130	135-140	150-155	165-170	180-185	195-200	210-215	225-230	240-545
	Hole No.	CC-74-2				·	*					

12/74-1/75 Rotary Drilling TD - 215'

							٠.								
		Ag OZ/T							.07	0.097	.07	0.097	.07	0.70	.07
(1)	ERC	일시								00	.02	0.016	19	100	00.
7	AS	Oxide Cu %							0.0	0.009	00	0.007	000	300	00.
		ದ್ದಿ							1.000		1.	0.200	100	. 22 . 22	.24
	Hawley & Hawley Assay	ದ್ದಿ		0.02	0.03	40.00	4.4	-,-	00	23	44	2.4	4	100	2
		Sample Interval		15-20	30-35	40-45	0 N N 1 N	0-6 5-7	5-2	5-9	0-95	00-	10-11	20-12	25-13
		Hole No.	CC-74-3												

12/74-1/75 Rotary Drilling TD - 215'

	Ag OZ/T	0.073	0.4	0	4	•	•	9	0	0	0	0	0	0
ERC Assay	110	0.015	00.	10.	.01	00.	00.	00.	00.	00.	00.	00.	00.	00.
Ψ.	Oxide Cu %	0.008	10.	.01	OTO.	00	.02	.03	LT.	11.	.07	-	4	· 07
	ಗ್ಗಿ	0.320	.21	23	.32	33	.09	4	.26	.23	.23	.27	.24	•16
Hawley & Hawley	n _c	0.34												
	Sample Interval	577	1-0-1	50-15	55-16	60-16	65-1	70-17	75-18	80-18	85-19	90-19	95-2	00-2
*	Hole No.	CC-74-3										•		

12/74-1/75 Rotary Drilling TD - 180'

	•1	7-74-D	15.	30	54	09	75	06		OT
Hawl	Sample <u>Interval</u>		15-20	30-35	45-50	60-65	75-80	90-95		105-110
Hawley & Hawley Assay	Cu %		0.01	0,01	0.01	0.01	0.01	0.15		0.26
	ng/						0.079	8440	0.089 0.145 0.245	.25
ERC	Oxide Cu %				* * *		0.014		0.022 0.024 0.028	•
C	0110						0.002	000	000000000000000000000000000000000000000	00
	Ag OZ/T						0.024		0.024	0.024
							(80-82.5°) (82.5-85°)	7 7	97.5-100') 100-102.5'	\

12/74-1/75 Rotary Drilling TD - 180'

				1) H		
		Hawley & Hawley Assay		As	ERC Assay		
Hole No.	Sample Interval	Ça Ç	n _w	Oxide Cu %	0 to	Ag OZ/T	
η-η ረ- ጋጋ			0.240	0.023	0.007	0.024	(110-112.5°) (112.5-115°)
	אמר טמר	C C	30	.02	000	0.0	(117.5-120')
	(21-021	•	0.530	461.0	0,000	420.0	127
			-7.	13-	00.	70	30-132
			2	.30	·OI	70.	32.5-13
	135-140	0.33	ښ١	0.05	00.	70	
			 	000	000	700	142
			4 -	0		70	45-147
			4	00	00.	70.	47.5-15
	150-155	0.17	٦,	00	10.	0.0	1
			٦,	00.	10.	.03	757
			4 -	. 0	- 0	200	01-7-75
			• 0				60 K-1 6
	165-170	0.20	14		30.	.07	01-0-20

12/74-1/75 Rotary Drilling TD - 245'

	AE OZ/T								
ERC	Oxide Ho							d .	
	ng Ca								
Hawley & Hawley	Ca Ca	0.01	0.01	0.03	£0°0	70°0	0.08	0.01	0.01
	Sample Interval	15-20	30-35	45-50	60-65	75-80	36-06	105-110	120-125
	Hole No.								

12/74-1/75 Rotary Drilling TD -, 245'

	AS OZ/T								
ERC	Oxide Mo								,
y	Ca								
Hawley & Hawley Assay	Ca	0.01	0.01	0.01	0.01	0.03	0.02	0.03	0.03
	Sample Interval	135-140	150-155	165-170	180-185	195-200	210-215	225-230	240-245
	Hole No.	CC-74-5				*			,

12/74-1/75 Rotary Drilling TD - 278'

			×	(25-27.51))	(35-37.5")	1-0-0	2.5-4		0-52.	(52.5-55)	5-57.	7.5-6	5-67	(67.5-70")	0-72.	2.5-7	((80-82,51)	2.5-8	
		Ag oz/T		0.050	.06	90.	0,0	10	.05	.07	60.	.07	0,0	00	10	.07	.05	0.	0.5	.05	
ERC	E S		0.003	10.	10.	10	01	.00	OI.	.02	10.	96	100	00.	00.	00.	00.	10°	00.		
	Oxide Cu %		0.005	.02	.02	10	10.	.01	.03	.03	.03	20,5	020	.02	03	.02	.01	.02	.02		
		n Co		0.029	.21	.22	0.0	127	.25	35	.32	.30	623	200	.29	.35	.21	.31	.27	.24	
	.ey														•						
	Hawley & Hawley Assay	ಬ್ರೀಟ	0.01		0.20				0.26					•			activities and activities	0.30			
		Sample Interval	15-20	7.	30-35	18			45-50				79 09					75-80			

9-42-22

12/74-1/75 Rotary Drilling TD - 278'

		85-87.5") 87.5-90") 95-97.5") 97.5-100")							
	Ag . OZ/T	000000							
ERC Assa <u>y</u>	86	000000000000000000000000000000000000000							
AS	Oxide Cu %	0.020 0.027 0.014 0.026							
	262 262	000000000000000000000000000000000000000							
Hawley & Hawley Assay	ng ca	0.29	40.0	90.0	0,16	0.13	0.14	0.13	90.0
Haw	Sample Interval	. 56-06	105-110	120-125	135-140	150-155	165-170	180-185	195-200
	Hole No.	9-74-22			*				

12/74-1/75 Rotary Drilling TD - 278'

ERC ASS21	Oxide Ho		
	og Ca		
Hawley & Hawley Assay	3 63 C	0.03	
	Sample Interval	210-215	
	Hole No.	9-44-22	

40.0

225-230

70°0

540-545

0.03

255-260

Ag OZ/T

ERC Assay	Oxide Ho	
	n Co	
Hawley & Hawley Assay	Cu %	<.01
	Sample Interval	5-10
	Hole No.	CC-74-7

<.01

20-25

.02

95-100

<.o1

110-115

50-55

35-40

65-70

80-85

12/74-1/75 Rotary Drilling TD -.165'

	Ag OZ/T			
ERC	Oxide Mo			
	ng.			
Hawley & Hawley Assay	Cu.	<.01	<.01	10
	Sample Interval	125-130	541-041	1 44-1 60
	Hole No.	2-42-22		,

12/74-1/75 Rotary Drilling TD - 65'

	Ag oz/T	0.0000 0.122 0.122 0.122 0.122 0.122
ERC	0 2	0.015 0.022 0.0048 0.004 0.012
As	Oxide Cu %	0.016 0.016 0.019 0.080
	ದ್ದಿ	00000000000000000000000000000000000000
Hawley & Hawley	n _W	0.03
	Sample Interval	2011 1011 1010 11110 1000 1000 1000 100
	Hole Wo.	CC-74-8A

		Ag OZ/T			0.109	4,4	0.	12	9 1 1 0	900	11.88	0.07	90.	000	.07	60.
9	ERCASSAN	170			0.020	10.	020	0.0	0.0	9.	10.	10.	.02	10.	02	.02
i 구	AS	Oxide Cu %			0.006	.07	0.00	.03	0	.02	10.	000	00	85	00	.01
		n d			0.0000	200	40.7	47	いなが	34	.26	200	18	.21	1.9	.23
	Hawley & Hawley Assay	n _{t2}		0.01	46.0		0.35	١	0.53	0.37		0.18		0.21		0.22
		Sample Interval		15-20	25-30	700	20	70	0 1V	07	010	5-10	00-10	$\mathcal{N}_{\mathcal{C}}$	15-12	20-12
		0 0 0	CC-74-8					21								

12/74-1/75 Rotary Drilling TD - 160'

	Ag OZ/T	00000000000000000000000000000000000000
ERCASSAY	0 KJ	0.023 0.039 0.038
AS	Oxide Cu %	0.009 0.007 0.006 0.011 0.019
	ನ್ನಿ	0.240 0.240 0.220 0.450
Hawley & Hawley Assay	Ca	44.0
	Sample Interval	1255 1355 1355 1455 1455 1550 1550
	Hole No.	CC-74-8

12/74-1/75 Rotary Drilling TD - 270'

	,	Ag oz/T									0.036
•	ERCASSAY	51 1	×								0.024
	ASS	Oxide Cu %		1						¥	0.012
		5 kg									460.0
	Hawley & Hawley Assay		10	0)	1	02	03	8		8	8
	y & F	ng C	0.05	0.02	0.01	0.02	0.12	0.03	0.01	0.03	0.03
	Hawle										,
		le rval		70	0	٧٠	0	70	5	110	125
		Sample Interval	5-10	20-25	35-40	50-55	02-59	80-85	90-95	105-110	120-125 125-130
		No	6-1				N)				
		Hole No.	6-44-00								

12/74-1/75 Rotary Drilling TD - 270'

		Ag OZ/T	00000000000000000000000000000000000000	540.0	* *					
)	ERCASSAY	ES SS	0.020 0.035 0.035 0.020	0.018						
1	AS	Oxide Cu %	00000	00					• ,	*
		ng Ca	00000000000000000000000000000000000000	0.195						
	/ley									
	Hawley & Hawley Assay	ng co	0.27	0.10	0.11	0.14	0.36	0.16	£0°0	20.07
		Sample Interval	1330 1330 1347 1140 1450 1547 1550	10100	180-185	195-200	205-210	220-225	235-240	250-255
		Hole No.	6-44-00							

12/74-1/75 Rotary Drilling TD - 265'

	Ag oz/T			200	000000000000000000000000000000000000000	0000		
ERC Assay					000000000000000000000000000000000000000			
	Oxide Cu %			604	0.0000	0400 5220		
	ng/			0 41	00000000000000000000000000000000000000	と よっ かん かん		
Hawley & Hawley Assay	250	0.10	0.08	0 4	00 00 00 00	0.45 0.27 0.28 0.16	0.10	0.12
	Sample <u>Interval</u>	5-10	20-25	447	0000	00000 00000 00000 11111 100000	105-110	120-125
	Hole No.	CC-74-10			8			

12/74-1/75 Rotary Drilling TD .- 265

Assay

Oxide Cu %

36

Hawley & Hawley Assay 25/ Sample Interval Hole No.

00-74-10

0,16 150-155

0.13 165-170 60.0

180-185

60.0

195-200

210-215

0.12

0.10

225-230

0.10

240-245

0.10

255-260

		AE OZ/T			0.061	0.00	0.055	0.061	0.075	
	ERCASSAY	07.7%	,		700.0	0.00%	0.002	0.002	0.003	
1	ASS	Oxide Cu %			•	0.036				
		ng S			α	0.160	$H \circ$	NH	\vdash	
	Hawley & Hawley Assay	ನ್ಯ		60.0	70	0 0 0		77.0	0.16	0.08
		Sample Interval		10-15	20-25	30-35	35-40	45-45	50-55	70-75
		Hole No.	CC-74-11							

COPPER CREEK PROJECT

CC74 SERIES ROTARY PERCUSSION HOLES (OCCASIONAL SECTIONS CORED)

DRILL CONTRACTOR - HARRIS - WOODIS (HUGH M. HARRIS DRILLING CO.)

PROJECT WORK PERFORMED: 12/74-1/75 4 3/4" Hammer/NQ Core

ABBREVIATED DESCRIPTIVE LOGS (See strip logs - attached - for relationship/rock types to assays)

CC-74-1 - TD 250 1/2

0-5 Fill

Quartz monzonite porphyry (Qmp); light gray except medium gray 115-125 and 150-160 (more mafic sections); medium grained except for granophyric texture 70-75 and 175-190.

Secondary biotite 90-160, including some phylogopite veinlets 125-130; heavy sections of quartz veinlets to 1/16" are common. Fe mags generally destroyed; oxidized to 85 w/medium to light brown limonite.

Trace to minor chalc. begins @ 40, moderate to heavy chalc. 85-200 (200-215 some covellite).

Trace cpy 145-200. MoS₂ prevalent 160-200. Pyrite < 1/2% 40-70; 1/2+% 70-90, 2-3% 90-130, less 130-170; ±2% 175-200. Qtz-py veinlets 110-115, turquoise 80-85.

- 200-205 Quartz latite porphyry (Qlp), variable, light pinkish gray; a fine grained porphyry some sericite. Minor to moderate chalc. Fine grained Py 1-2%.
- 205-250 1/2 Quartz monzonite porphyry (Qmp) as above 200' but somewhat granophyric 225-250 1/2, variable light to medium gray, siliceous; no secondary biotite noted.

Minor to moderate chalc.; trace to minor cpy; trace MoS_2 ; Py 1-2%.

Water table @ 245' (cored 247'1/2 - 250 1/2)

CC-74-2 - TD 250 1/2

0-5 Fill

- 5-35 Quartz monzonite (Qm) medium grained; oxidized; buff to orange in color; Fe oxide stringers to 35'.
- 35-222 1/2 Quartz latite porphyry (Qlp or Quartz monzonite porphyry (Qmp); variable light green gray to medium gray; silicified (locally heavy). Fe mags well chloritized.

Trace to moderate chalc., variable cpy (more than in most holes); trace to minor NoS2, increasing w/qtz veinlets 205-210; 115-125 quartz veinlets carry Py, NoS2, trace cpy. Py ±2%.

Water table @ 140'. (cored 140-149.2). Some dilution of sample below 165'.

- 222 1/2-225 <u>Latite-Rhyolite</u> dike post ore; buff colored, only minor pyrite.
- 225-235 Quartz latite porphyry Quartz monzonite porphyry; light green gray, w/some flow banded rhyolite. Qlp-Qmp is siliceous (then bone white).

2-3% Py, trace to minor Cpy and MoS2.

235-250 1/2 Quartz latite (Q1) - Rhyolite (Rhy), light green gray; chloritized ferromags; w/abundant quartz veinlets.

1-2% Py; some cpy; trace to minor MoS2

CC-74-3 - TD 215

0-5 Fill

Quartz latite porphyry (Qlp) - Quartz monzonite

porphyry (Qmp) medium gray to dark pinkish gray;
a mafic Qlp-Qmp below 30'; oxidized - leached to
30' - w/heavy Fe oxide fracture coatings. Possible
secondary biotite 55-60.

Sulfides begin @ 35'; minor chalc; trace cpy, 2-3% Py.

- 65-70 Quartz monzonite porphyry (Qmp) dike medium green gray; propylitized (chloritized Fe mags, epidotes, hematite) a little FeS2.
- 70-160 Quartz monzonite porphyry (Qmp) mafic; dark green gray, fine grained; brown shredded secondary biotite 90' and below; minor to some chalc., but

heavy 70-75; trace to some cpy; trace MoS2 145-150; 1-2% Py (tarnished).

- Quartz latite porphyry (QIp) leucocratic; buff; w/well bleached Fe mags; trace chalc; trace cpy and minor Py.
- Monzonite (Monz); equigranular; fine grained; buff; w/some hairline qtz veinlets; some chalc; generally low pyrite.

CC-74-4 - TD 180

0-5 Fill

- 5-35 Quartz-Feldspar Rock (QFR) (a well oxidized Qmp?); orange; bleached biotite, some quartz veining; heavy Fe oxides after Py.
- 35-180 Quartz monzonite porphyry (Qmp); medium light gray to buff; silicified; bleached biotite to 170'. Biotite 70% fresh below 170'; abundant Fe oxides to 85-95. Bottom zone of oxidation @ 100'.

Some Kspar flooding 160-170.

CC variable, locally heavy; trace cpy; MoS₂ w/quartz veinlets 150-155, Py diminishes to +1% below 115'.

CC-74-5 - TD 245

0-5 Fill

Quartz monzonite (Qm) medium grained, light yellow to yellow gray; well oxidized to 60' w/abundant limonite coatings; w/possible quartz latite (silicified) 55-60 (dike?); quartz veinlets are prevelant below 60'.

 $\pm 1/2-2\%$ Py to 60'; trace chalc and cpy 35-60'. Below 60' - Py $\pm 1\%$; very minor chalc, trace cpy and minor MoS2.

- 85-105 <u>Latite</u> (Lat); fine grained; light gray; silicified; 1/2+/2 Py; trace chalc.
- Quartz monzonite porphyry (Qmp); as above 85'.

 Light pink gray to green gray darker toward bottom; no Fe mags noted 140-190; elsewhere biotite generally bleached; but possibly secondary biotite. 235-245'.

Trace hematite 200-205, 1/2-1 1/2% Py, possible occasional trace chalc, trace cpy below 225; trace MoS₂ below 195'.

Water table @ 190'.

<u>cc-74-6</u> -	TD 278
0-5	Fill
5-15	Quartz-Feldspar Rock (QFR) yellow-brown; well leached - oxidized; Fe (Mn?) coatings well developed; minor silicification; no sulfides noted.
15-20	Latite (Lat) green gray - oxidized to orange; (other comments as above).
20-65	Quartz monzonite porphyry (Qmp) - Quartz latite porphyry (Qlp) medium to dark gray.
	Some chalc; trace cpy; 1-2% Py except +3% Py 30-35.
65-95	Biotite Quartz Monzonite (Bqm); medium to dark gray.
	Some chalc; trace MoS_2 90-95. Py 2-3% - including with biotite in veinlets.
95-100	Quartz latite (Q1) dike - very light gray, w/numerous quartz veinlets; only minor chalc; 2-3% Py.
100-105	Quartz latite porphyry (Qlp) - quartz monzonite porphyry (Qmp); medium gray; with some 1/8" quartz-Py-MoS ₂ veinlets, minor chalc.
105-170	<u>Latite-rhyolite</u> (Lat-Rhy); light to medium gray; trace to minor MoS ₂ ; trace chalc; <u>+</u> 2% Py below 110.
170-278	Quartz monzonite porphyry (Qmp); light gray 170- 175; medium to dark gray 175-278; fine to medium grained with sucrosic texture 180-215; bleached biotite;
	1-2% Py below 180' (2-3% - 220-260'). Trace to very minor chalc, trace MoS2; rock almost devoid of sulfides 70-180'.

Water table @ 245'. (cored NQ - 272 1/2 - 278)

CC-74-7 - TD 165 0-5 Fill Quartz monzonite porphyry (Qmp); buff to orange; 5-35 medium grained; well oxidized - leached; w/very few remaining Fe mags; Fe-Mn oxide coatings; some silicification. Quartz latite porphyry (Qlp); medium gray; medium 35-165 to coarse grained; some bleached biotite, possible sericite, trace tarnished Py. CC-74-8 - TD 160 Monzonite; fine grained, yellow brown, leached; 0-40 good Fe oxide coatings below 32 1/2, green gray, fine grained; w/possible secondary biotite. Good chalc; trace cpy; +3% Py including quartz pyrite veinlets. Mixed Monzonite (monz) fine grained and Quartz 40-45 latite (foliated) - light green gray. good chalc; trace cpy, +3% Py. Quartz latite (Q1) light pink gray -- otherwise as 45-55 40-45 but with more cpy. Mixed Q1-Monz - (as 40-45) Q1 dikes cut monz. 55-65 green gray w/some dark gray mafic dikes; good chalc; possible trace native Cu; trace to minor cpy, some Py. Quartz latite porphyry (Qlp) light pink to green 65-70 gray (translucent) w/quartz veinlets (w/Py-cpy) good chalc. Mixed Olp-Om (fg Biotitic Qm); variable yellow 70-150 white to dark green - motted. Qlp predominating -Qlp cuts Qm as 55-65; possible secondary biotite below 120. Chalc variable; cpy - trace to moderate amounts; trace to some MoS₂ 115-160; Py variable 1-2%; possible native Cu 130-135. Quartz monzonite (biotitic); mottled blue green 150-160 w/dark brown biotite-Py veinlets; some quartz

veinlets; substantial cpy; trace MoS2; +3% Py.

CC-74-8A - TD 65

0-20 Soil and gravel

Quartz latite (Q1) - Quartz monzonite (Qm); medium green gray; w/granophyric textures; good chalc; trace cpy - MoS2; 2-3% Py

Water table @ 40'.

40-42 1/2 No sample.

42 1/2-65 (cored) Mixed Quartz Monzonite porphyry (Qmp) and mafic dikes; Qmp green gray, well fractured, with chloritized hornblende fracture fillings; abundant quartz veinlets > 1/4" w/Py, cpy, MoS2. 5-10% total sulfides w/some chalc.

CC-74-9 - TD 270

0-5 Fill

Biotite quartz monzonite porphyry (Bqmp) light to medium gray and off white, locally green gray (mottled). Biotite destroyed to 50'; leached, oxidized w/Fe oxides to 50'; some oxidation continues to 120'.

Trace Py 50-100, 1-2% below 190, 2-3% 190-230. Chalc begins at about 100, increases @ 125, less to none below 160'; cpy variable, trace 135-210, minor 210-230; MoS_2 in trace amounts 150-230.

Water table @ 205'.

- 230-240 Quartz monzonite porphyry (Qmp) dike green gray; w/shredded chlorite; very little sulfide.
- 240-270 "Crowded" biotite quartz monzonite porphyry (Eqmp) med gray rock w/dark brown biotite books including secondary biotite; 1-2% Py, trace cpy and MoS2.

CC-74-10 - TD 265

O-225 Quartz monzonite (Qm) fine to medium grained; mafic; mottled dark green to green gray; heavily propylitized (30% epidote, chlorite quartz); w/possible mafic dike 150-155; well oxidized to 50 w/abundant Fe oxides.

Some quartz veinlets; increasing to 20-30% of rock as quartz veinlets 140-150, 10% 160-225. Some CaCo3 veinlets 0 215.

Trace chalc; trace to minor cpy; trace MoS2; Py - 1-3%.

- 225-235 Fault gouge and mafic quartz monzonite
- 235-265 Om as above 225 but more biotitic w/some zones of Olp. Trace cpy, no visible chalc.

CC-74-11 - TD 85

- 0-15 Quartz monzonite? (Qm) heavily oxidized leached; good Fe oxides.
- 15-45 Quartz latite (Q1) (foliate); mottled dark green to green gray; heavily chloritized. 2-3% Py, trace to minor cpy and chalc.
- 45-50 Quartz monzonite (dike) <u>leucocratic</u> pink to tan minoc chalc.
- 50-85 Quartz latite (Q1) mafic, mottled color as 15-45, heavily chloritized; 2-3% Py, trace cpy, trace to minor MoS2; no visible chalc.



ROCKY MOUNTAIN GEOGHEMICAL CORP.

2561 EAST FORT LOWELL ROAD . TUCSON, ARIZONA 85716 . PHONE: (602) 795-9780

Certificate of Analysis

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March 14, 1975

Client:

Earth Resources P. O. Eoz 698

Cuba, New Mexico

87013

2. Page 1 of

RMGC Numbers:

Local Job No.: 75-4-36TC

Foreign Job No.:....

Invoice No.: S 5511

Client Order No.:

Report On:

3 of 15 samples

Submitted by:

E.E. Saffell

Date Received:

liarch 10, 1975

Analysis:

Qualitative spectrographic analysis.

Analytical Methods:

Remarks:

Percent ranges given are approximate.

cc:

Inc.

A. Perry

J.N. Swinderman

RMCC: SLC

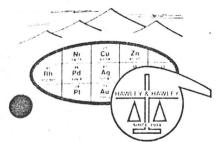
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PDW/sl

Client	Earth Resources	Date Max	rch 14, 1975 RMGC Job	No. 75-4-3670
			F	Page 2 of 2
	Sample No.	(+1%) <u>Najor</u>	(.01% to 1%) <u>Minor</u>	(less than .019
	12	Aluminum Iron Silicon Sodium Titanium	Calcium Copper Magnesium Molybdenum(low) Potassium	Nickel Silver Strontium
	Approprieta de la companya del companya de la companya del companya de la companya del la companya de la compan	nder solder film so, in 20 prins sign wager of open name and and any overlands in 2000 called a policy of the contract of the		Venadium Zinc
	Sample No.	Major	Minor	Trace
	13	Aluminum Iron(low) Potassium Silicon Sodium	Calcium Copper Nagnesium	Barium Lead Manganese Molybdenum Nickel Silver (low) Titanium Zinc
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7.4	Sample No.	Major	Minor	Trace
	14	Aluminum Tron(low) Silicon Sodium	Calcium Copper Magnesium Potassium Titanium	Barium Lead Manganese Molybdenum Nickel Silver(low) Vanacium Zinc

By hours D. Willand





SKYL LABS, INC.

Hawley &wley, Assayers and Chemists Division 1700 W. Grant Rd., P.O. Box 50106, Tucson, Arizona 85703 (602) 622-4836



Charles E. Thompson Arizona Registered Assayer No. 9427

William L. Lehmbeck Arizona Registered Assayer No. 9425

CERTIFICATE OF ANALYSIS

NO.	SAMPLE IDENTIFICATION	Cu %	N.S.Cu %	N.S.Cu %						
			*	**	ē					
						is .				
1	Composite: CC-74-10	0.41	0.24	0.27						
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P.O. Box 12754 Tucson, Arizona 85732

Attn.: Mr. A. J. Perry

DATE REC'D: 1/29/75 DATE COMPL.: 2/13/75 JOB NUMBER: 750132-A

GRADE CALCULATION IN COPPER EQUIVALENTS

,		NORTH	AREA		d 0	
Hole No.	Total Depth	Ore Intervals	% Cu (Total)	% MoS2	% Cu as <u>Mo eq</u>	% Cu equivalent
EH-4	165	@60'-35'	0.24			0.24
71-8	480	@90'-120'	0.36	.02	•06	0.42
74-9	270	@130'-30'	0.27	.016	.05	0.32
74-8A	65	@20'-15'	0.36	.008	.03	0.39
74-8	160	@30'-125'	0.33	.02	.06	0.39
Ен-6	245	@30'-35'	0.31			0.31
74-1	$250\frac{1}{2}$	@80'-165'	0.58	.018	.06	0.64
EH-1	175	@140'-35'	0.37			0.37
EH-2	205	@90'-65' @185'-20'	0.26 0.50			0.26 0.50
CC-3	886 .	@110'-185'	0.38	.0115	.04	0.42
74-2	$250\frac{1}{2}$	@45'-95'	0.24	.02	.06	0.30
74-10	265	@55'-35'	0.45	.006	.02	0.47
EH-3	165	@20'-25' @90'-75'	0.34 0.50			0.34 0.50
74-3	215	@70'-35' @120'-45'	0.35 0.28	.012	.04 .03	0.39 0.31
74-11	85	@20'-10'	0.28	.003	0.1	0.29

OREBODY GRADE CALCULATION

NORTH AREA

Hole	Tons Ore (Total)	Tons Ore Detail	Grade % Cu (Cu eg)	Tons X Grade
EH-4	248,640		0.24	59,674
71-8	1,209,600		(0.42)	508,032
74-9	271,680		(0.32)	86,938
74-8A	39,840		(0.39)	15,538
74-8	540,000		(0.39)	210,600
ЕН-6	271,040		0.31	84,022
74-1	2,529,120		(0.64)	1,618,637
EH-1	208,320		0.37	77,078
EH-2	527,680	124,160 403,520	0.50 0.26	62,080 104,915
CC-3	2,829,760		(0.42)	1,188,499
74-2	2,036,800		(0.30)	611,040
74-10	579,040		(0.47)	272,149
EH-3	764,800	573,600 191,200	0.50	286,800 65,008
74-3	824,320	360,640 463,680	(0.39) (0.31)	140,650 143,741
74-11	112,640	* * * *	(0.29)	32,666
Total:	12,993,280		Total:	5,568,157

GRADE = % Tons = 0.428% Cu eq.

NORTH AREA OREBODY

e Reserve Calculation

Hole No.	Area of Influence ft ²	Thickness of Ore ft	Volume of Ore ft3	Tons Ore (12.5 ft ³ /ton)	Tons Ore Detail
EH-4	. 88,800	35	3,108,000	248,640	
71-8	126,000	120	15,120,000	1,209,600	
74-9	113,200	30	3,396,000	271,680	
74-8A	33,200	15	498,000	39,840	
74-8	54,000	125	6,750,000	540,000	
ЕН-6	96,800	35	3,388,000	271,040	
74-1	191,600	165	31,614,000	2,529,120	
EH-1	74,400	35	2,604,000	208,320	
EH-2	77,600	20 65	1,552,000 5,044,000	527,680	124,160 403,520
cc-3	191,200	185	35,372,000	2,829,760	
74-2	268,000	95	25,460,000	2,036,800	
74-10	206,800	35	7,238,000	579,040	
EH-3	95,600	75 25	7,170,000 2,390,000	764,800	573,600 191,200
74-3	128,800	35 45	4,508,000 5,796,000	824,320	360,640 463,680
74-11	140,800	10	1,408,000	112,640	
			Total Tons Ore:	12,993,280	

NORTH AREA OREBODY

Overburden Tonnage Calculation

				•												
Total Tons Waste	426,240	907,200	1,177,280	53,120	129,600	232,320	1,226,240	833,280	744,960	1,682,560	964,800	909,920	497,120	875,840	225,280	
Tons Internal Waste									155,200				344,160,	154,560		•
Internal Waste Volume ft								30 40 3	1,940,000	w w		*	4,302,000	1,932,000		
Internal Waste Thickness ft	*							:.	20.				. 45	15		
Tons Overburden	426,240	907,200	1,177,280	53,120	129,600	232,320	1,226,240	833,280	589,760	1,682,560	964,800	909,920	152,960	721,280	225,280	
Overburden Volume ft3	5,238,000	11,340,000	14,716,000	000,499	1,620,000	2,904,000	15,328,000	10,416,000	7,372,000	21,032,000	12,060,000	11,374,000	1,912,000	9,016,000	2,816,000	
Overburden Thickness ft	09	06	130	20	30	30	80	140	95.	110	45	55	50	70	20	
Hole No.	EH-4	71-8	6-42	74-8A	74-8	EH-6	74-1	EH-1	EH-2	6-00	74-2	74-10	ЕН-3	74-3	74-11	

Total tons Overburden: 10,885,760

GRADE CALCULATION IN COPPER EQUIVALENTS

SOUTH AREA

Hole No.	Total Depth	Ore Intervals	% Cu (Total)	% MoS ₂	% Cu as Mo eq	% Cu equivalent
74-4	180	@105'-35'	0.37	.006	.02	0.39
7.0-1	1500	@20'-85'	0.27			0.27
PD-1	607	@3'-150'	0.43			0.43
71-5	401	@50'-35'	0.38			0.38
на-6	265	@65'-160'	0.41.	.02	.06	0.47
'HA-11	300	@35'-75'	0.48			0.48
74-6	278	@45'-50'	0.28	.008	.03	0.31
70-2	581	@50'-125'	0.32			0.32

C. BODY GRADE CALCULATION

SOUTH AREA

Hole No.	Tons Ore_	Grade % Cu (Cu eg)	Tons X Grade
74-4.	411,040	(0.39)	160,306
70-1	671,840	0.27	181,397
PD-1	1,785,600	0.43	767,808
71-5	399,840	0.38	151,939
на-6	2,534,400	(0.47)	1,191,168
HA-11	1,032,000	0.48	495,360
74-6	748,800	(0.31)	232,128
70-2	788,000	0.32	252,160
Total:	8,371,520	Total:	3,432,266

GRADE = % Tons = 0.410% Cu eq.

SOUTH AREA

NO.	urden	,120	158,080	35,712	571,200 (009,	481,600	673,920	315,200	,432
CULATION	Tons Overburden	1,233,120	158	35	571	1,029,600	481,	673,	315,	Total: 4,498,432
OVERBURDEN TONNAGE CALCULATION	Overburden Volume	15,414,000	1,976,000	446,400	7,140,000	12,870,000	6,020,000	8,424,000	3,940,000	Total:
OVERBURD	Over- burden Thickness	105	50		20	65	35	45	50	
	Tons	411,040	671,840	1,785,600	399,840	2,534,400	1,032,000	748,800	788,000	8,371,520
TION	Volume of Ore ft	5,138,000	839,800	22,320,000	4,998,000	31,680,000	12,900,000	9,360,000	9,850,000	Total:
OREBODY TONNAGE CALCULATION	Thickness of Ore	35	85	150	35	160	75	50.	125	
OREBODY TON	Area of Influence ft2	146,800	98,800	148,800	142,800	198,000	172,000	187,200	78,800	
	Hole No.	4-46	70-1	PD-1	71-5	HA-6	HA-11	9-72	70-2	

PIT W.L LAYBACK WASTE CALCULATION

	Area (ft ²)	Ave. Depth (ft)	Volume	Tons	½Tons	Over- burden/ Layback
NORTH	538,400	127	68,376,800	5,470,144	2,735,072	3.98
SOUTH	396,800	92.5	36,704,000	2,936,320	1,468,160	3.06
		Pi	t wall layba	ck waste:	4,203,232	

STRIP RATIO CALCULATION

	Tons Over- burden	Tons Pit wall Layback	Total Tons Waste	Tons Ore	Waste/ Ore
NORTH	10,885,760	2,735,072	13,620,832	12,993,280	1.05/1
SOUTH	4,498,432	1,468,160	5,966,592	8,371,520	0.713/1
Total:	15,384,192	4,203,232	19,587,424	21,364,800	0.917/1

August 7, 1974

To: W. C. Cole

From: J. N. Swinderman

Re: COPPER CREEK PROJECT-PROPOSED EXPLORATION & DEVELOPMENT

We have thus far completed 2 diamond drill holes this year at Copper Creek, both of which penetrated extensive sections of altered and mineralized volcanic textured porphyries before bottoming in very weakly mineralized Precambrian rocks. Although assays are somewhat disappointing, we have tested less than half the intrusive complex and have yet to penetrate the intrusive center. Our holes indicate a plunge to the west. Deep potential is still good for several hundred million tons of 0.7% Cu equivalent if we can find the intrusive center.

Hole CC-3 penetrated 135 feet of 0.42% Cu and 0.01% Mo at a depth of 135 feet. This mineralization represents a shallow enrichment blanket near the eastern contact of the intrusive. Similar mineralization was encountered in EH-1, EH-2 and EH-3. Excellent open pit potential exists for about 12 million tons of 0.5% Cu - .025% Mo at about a 1:1 - waste: ore ratio.

Several holes along the southern contact of the intrusive indicate enriched copper values in brecciated and sheared rocks near ore grade (holes PD-1, HA-6, HA-11, DDH-70-2). Potential there looks fair for an aggregate of some 15 million tons of 0.4 - 0.5% Cu - 0.02% Mo at less than a 1:1 waste: ore ratio.

I recommend that we test the shallow enriched areas by means of rotary drilling (ll holes, average depth 250') and test the deep potential with one more diamond drill hole to a possible depth of 3,000'.

Our recent drilling has essentially substantiated previous interpretations that we are dealing with a major porphyry copper sulfide system, but primary grades are low and we have been unable to recognize a distinct grade increase with depth. Our drill holes have not tested the hypothesized parent quartz monzonite intrusive because both holes drilled through the complex into Precambrian rocks. The intrusive complex is either plunging to the west or has silled out into the Precambrian from feeder dikes. The lack of mineralization below the complex compared to that in hole PD-6 indicates the former. Considering the potential target, I still believe we have a good exploration bet.

W. C. Cole Copper Creek Project oposed Exploration & Developm J. N. Swinderman 8/7/74 Page Two

Monthly payments of \$1,100 will start September 7, 1974 for the Trust 60 ground.

Assay summaries of our drill holes, a generalized geologic and drill hole map, and a conceptual cross section are enclosed, please refer to logs prepared by Al Perry for detailed rock descriptions.

Budget Estimate: To Complete Rec	commended Progra	<u>m</u>
Core Drilling 3000'	@\$16.00	\$ 48,000
Rotary Drilling 3000'	@\$5. 00	15,000
Mobilization-Demobilization	@\$1,000	1,000
Roads	@\$5,000	5,000
Consulting 30 days	@\$110	3,300
Air travel 5 trips	@\$125	625
Auto 5 trips	@\$80	400
Meals, lodging, entertainment (5)	@\$50	250
Labor 20 days	@\$40	. 800
Assays 750 Cu-Mo	@\$4.50	3,375
Land - Gillings 4 months	@\$550	1,650
Trust 60 4 months	@\$1,100	4,400
Miscellaneous		1,200
		\$ 85,000

W. C. Cole Copper Creek Project-Proposed Exploration & Development J. N. Swinderman 8/7/74 Page Three

ASSAY SUMMARY

CC-3

	Interval	Intercept	0/0 Cu	0/0 Mo	Equiv. @ Mo.=3.2xCu		
	0 - 9	9'	No Core	No Core			
	9 - 110	101'	0.04	0.01	0.07		
	110 - 135	25 '	0.24	0.009	0.27		
	135 - 270	135'	0.42	0.0115	0.46		
ř	270 - 445	175'	0.28	0.0247	0.36		
	445 - 690	245'	0.20	0.0145	0.25		
	690 - 886	196'	0.12	0.003	0.13		
CC-4							
	12 - 160	148'	0.18	0.006	0.20		
	160 - 440	2801	0.10	0.004	0.11		
	440 - 1030	590'	0.15	0.02	0.22		
	1030 - 1120	90'	0.26	0.03	0.36		
	1120 - 1380	260'	0.13	0.01	0.16		
	1380 - 1420	40'	0.08	0.03	0.16		
	1420 - 1470	50 '	0.01	0.001	0.01		
	1470 - 1611	141'	0.04				

J. N. Swinderman

JNS:rm

January 31, 1974

To:

W. C. Cole

From:

J. N. Swinderman

Re:

LITTLE COPPER CREEK

SUMMARY and RECOMMENDATIONS:

Extensive efforts by various exploration companies have defined a major sulfide system at Little Copper Creek but have failed to define a distinct orebody. In spite of excessive drilling and negative interpretations, I believe the most promising area has yet to be tested. Field interpretation of outcrops, core and available data leads me to believe that we are looking at the upper, subvolcanic expression of a porphyry copper intrusive complex which crops out on land largely controlled by Earth Resources Company and Day Mines. I recommend that we test this area for economic mineralization by means of diamond drilling.

GEOLOGY and MINERALIZATION

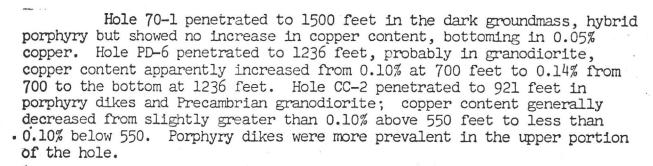
The area of the enclosed map depicted as quartz porphyry includes rocks with volcanic textures which have been intruded by complex, variable textured quartz monzonite porphyry "dikes". The "dikes" contain impressive stockwork veinlets of quartz, pyrite, chalcopyrite and molybdenite. Pyrite and chalcopyrite are disseminated in the groundmass. Chalcocite has replaced much of the fine chalcopyrite in surface outcrops. Alteration assemblages are argillic in the volcanios and clay-sericite to phyllic in the "dikes". Argillic and propylitic assemblages are present in the Precambrian rocks.

There is an excellent exploration chance that the quartz monzonite porphyry dikes are genetically related to a major porphyry copper sulfide system. If so, with depth, volcanic textures should give way to porphyry textures and argillic and clay-sericite mineral assemblages will give way to phyllic or biotite-orthoclase assemblages containing possible ore grade, primary mineralization.

DRILLING

Thus far, drilling has been concentrated in the Precambrian granodicrit. Splites and hybrid porphyry, where obvious oxide copper occurrences at fracture controlled.

W. C. Cole LITTLE COPPER CREEK J. N. Swinderman Page Two



Except for the shallow holes recently drilled by Essex, the porphyry has only one significant test, hole 70-3. Copper content increased from about 0.12% to about 0.18% in the quartz monzonite porphyry intercepted from 420 feet to the bottom of the hole at 590 feet. Stockwork veining and potassic alteration also increased markedly in this interval.

Holes HA-1, HA-9, EH-2, EH-3 and EH-19 encountered chalcocite enrichment near ore grade in the porphyry intercepts. This enrichment was apparently not bottomed in holes EH-2, EH-3 and EH-19.

LAND CONSIDERATIONS

In the area of interest land control is shared by four owners. The patented Dean Rose ground east of the intrusive center has an asking price of \$300,000. The owners will presently not consider an option agreement. Because of the price this land is not worth tying up at this stage. If an orebody is discovered we will have to negotiate for any needed ground.

The Trust 60 ground on the west edge of the intrusive center will go for a 3% NSR royalty, but owners are presently asking about \$3,600 front money and advance royalties of \$1200 per month for the first year, escalating to \$50,000 annual minimum after the fourth year. Because our drilling will benefit this ground, we might negotiate a free one year option if we drill the property ourselves. We should attempt this and also attempt to reduce advance royalty payments. If successful we should option this ground before drilling.

The Green Mountain Mill Site is owned by a woman who considers it of sentimental value and who is asking an unrealistic \$56,000 for the five acre site. This ground is not critical to exploration and need not be considered until a mine is developed.

Company are Day Mines and presently commands \$500 per month payment for the two parented Little Copper claims. If Days Mines considers the project worthwhile they should continue to share expenses. If not, we should finance the initial drilling ourselves.

W. S. Cole LITTLE COPPER CREEK J. N. Swinderman Page Three



Our objective is to test the target area by drilling. A vertical diamond drill hole located near the center of the most altered area should either substantiate or negate the concept of potential ore grade mineralization at depth and will further test the chalcocite enrichment encountered in hole EH-2. Some provision should be allowed for alternative drilling if unexpected conditions are encountered. Results of this drilling should be evaluated before additional work or abandonment is recommended. Positive results will require at least 4000 feet of additional drilling to assess the advisibility of developing the deposit or selling it to a major producer.

BUDGET

A Budget estimate based on one 1500 foot diamond drill hole follows:

Drilling 1500 NX min. Mobilization-demobilization Consulting 15 days Gillings payments 4 months Air travel 3 trips Auto 3 trips Meals, Lodging, Entertainment 3 trips Labor 7 days core splitting Core storage 4 months Miscellaneous Assay 70 Cu- Mo.		16 500 110 500 100 70 35 25 20 700 4	24,000 500 1,650 2,000 300 210 105 175 80 700 280
		-	\$30,000

Tr	rust 60 Lai 4 Months	Royalty			3600 1200	3,600 4,800
			*	*		\$38,400

J. N. Swinderman

JNS:rm

PERRY, KNOX, KAUFMAN, INC.

MINERAL EXPLORATION AND DEVELOPMENT

OFFICES:

TUCSON, ARIZONA (BUSINESS)

2343 E. BROADWAY, SUITE 206 P. O. BOX 12754, ZIP 85732 TELEPHONE (602) 622-0582

SPOKANE, WASHINGTON

NORTH 20 PINES ROAD. SUITE 21 P. O. BOX 14336. ZIP 99214 TELEPHONE (509) - WA 4-0878

Tucson, Arizona April 8, 1975

Mr. J. Bruce Imswiler Manager of Expl.-W.USA

IMC Suite 12, 390 Freeport Blvd. Sparks, Nevada 89431

As you know, Earth Resources Company and PKK recently completed shallow rotary percussion drilling at Copper Creek (Little Copper Creek) near Prescott -- in the process outlining a small reserve of low grade sulfide mineralization, considered sub-economic at this time. Earth Resources is not prepared to undertake additional exploration in search of deeper, primary ores. We now seek to option the properties under our control to a Company interested in searching for these possible deeper copper-molybdenum sulfide ores. As your Company and others have shown an interest in learning more of this opportunity we submit the attached data for your study.

Tabulation of Data:

- Summary Report, Near-Surface Copper-Molybdenum Sulfide Mineralization at Copper Creek, March, 1975.
- 2. Claim Map Scale 1"- 500'.
- 3. Generalized Geologic and Drill Hole Map Scale 1"-500'.
- 4. Conceptual Geologic Section 1"-500'.
- 5. Geologic-Assay log CC-4 (Earth-PKK's deep hole).
- 6. Recommendation of Swinderman Jan. 31, 1974.
- 7. " Aug. 7, 1974.

Other data including the logs of other Earth-PKK holes (CC1 thru CC3) and some Norandex holes; as well as all Earth-PKK cores and some Norandex cores are available for inspection to seriously interested parties.

The terms of option for both the Gillings and Trust 60 lands and terms proposed by Earth-PKK are outlined in the attached pages.

Earth-PKK expect to hold the Trust 60 ground for a reasonable period of time in order to allow evaluation by others to whom this data is submitted, but we urge: 1) your expeditious study if interested, or 2) your prompt rejection if the exploration concept, the risk, option terms, etc. are such as to be of no interest to your Company.

In the event your Company does not enter into an Agreement with ERC-PKK on the subject properties we request that all data provided be promptly returned to PKK-Tucson office.

We thank you for your initial interest. We would be pleased to answer any questions you might have.

A. J. Perry

PERRY, KNOX, KAUFMAN, INC.

copy: Earth Resources, Mr. Wm Cole

Attachments

ABSTRACT

Gillings Exploration and Option Agreement:

Effective date- July 14, 1971

Property - Little Copper #2 and #3, patented lode claims (see claims map)

Term - 5 years

Schedule of Payments - currently and until expiration of term, \$550/mo + \$300 interest each July 14.

Termination - upon 90 days written notice

Total purchase price - \$65,000 - all payments to apply (remaining to be paid, about \$48,000)

Trust 60 Lease Agreement and Option to Purchase

Effective date - March 7, 1974.

Property - 15 patented claims, 8 unpatented lodes and 2 unpatented millsites (see claims map)

Term - 20 years and thereafter, if mining

Schedule of Payments -

- 1) Advanced Royalties currently- \$1650/mo.

 March 7, 1976 thru Feburary 7, 1977 \$2200/mo.

 March 7, 1977 thru term \$4125/mo.
- 2) Production royalty 3% NSR thru buyout.

Termination - 30 days notice

Purchase Price - \$5 million (all payments apply)

PROPOSED TERMS TO PROSPECTIVE OPTIONEES - Copper Creek

- 1) Optionee would receive a 1 year (minimum) option assuming all property burdens (payments, assessments, etc.)
- 2) At the end of 1 year, if optionee desired to continue, a cash payment of \$100,000 would be made Earth Resources; with optionee naturally continuing with all obligations during ensuing option months.
- 3) If optionee desired to continue beyond the end of the 3d year, a further payment of \$500,000 would be made ERC-PKK. Earth and PKK would have no further interest in the property or the project.
- 4) After the first year, termination could be achieved with 30 days notice.

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microscope examination of wet CC74 series cuttings.

Silicification is particularly well-developed in the North Area -- perhaps to the extent of making a substantial difference in any anticipated excavation.

The obvious overlapping ("telescoping") of alteration types was early recognized by Sayers of PKK and supported by the limited petrographic work accomplished.

RESERVE SUMMARY

Reserves of copper/molybdenum-bearing material available in the two areas outlined as North and South areas on Figures 2 and 3, using a 0.25% Cu eq. exterior cutoff (0.20% Cu eq. interior cutoff) are as follows:

North Area - 12,993,280 tons - 0.428 Cu eq.

South Area - 8,371,520 tons - 0.410 Cu eq.

Grand Total- 21,364,800 tons - 0.42 Cu eq.

Total waste measured, both areas, was 19,587,424 tons, resulting in a waste (all categories) to ore ratio of 0.92:1.

You are referred to the appendix for the detail and further summary of the reserve and waste calculations.

OREBODY CONFIGURATIONS

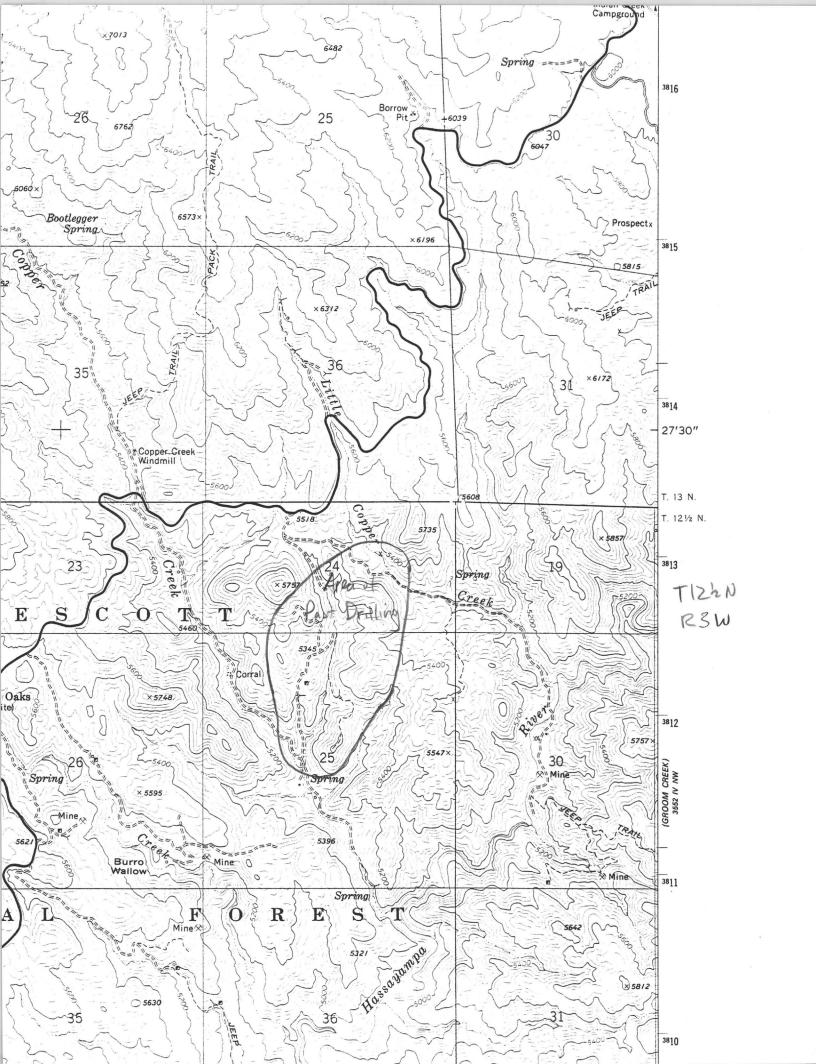
Figure 2, used as an overlay to Figure 1, will show that the North Area "orebody" has dimensions of approximately 1800' X 1100'. Ore thicknesses vary from 10 to 185 feet. Sections A, B, C and D show that the ore is generally flat lying and irregularly tabular, with a slight westerly dip. The ore (using a 0.25% Cu eq. cutoff) tends to be thicker on the west side of the body. There is a tendency for the better zone to drape downward with decreased surface elevations. Chalcocite was measured in the top of DDH CC-4 located just outside the west edge of North Area oreblock (but inside the proposed pit), illustrating that chalcocite can extend beneath valley bottoms, though this may be an exception.

The South Area (use overlay, Figure 3) is an irregular ellipse measuring approximately 1600' X 700' in plan. Ore thicknesses vary from 35 to 160 feet. Sections E, F, G and H illustrate interpreted ore configurations. A particularly good thickness of chalcocite is measured in HA-6, which was collared

on a topographic prominence.

3/4/92 - Brief Review suggests Hot the serve in both NES awas is you for Alfer Street Review suggests Hot the serve in both NES awas is you for Authority, etc. Pulling expension laterally, while not it depth, perhaps so wil lower cut-off, etc. Pulling street laterally which was to hollow chalcoute. Need to likely vivited to finding deep sulfield with the shocky.

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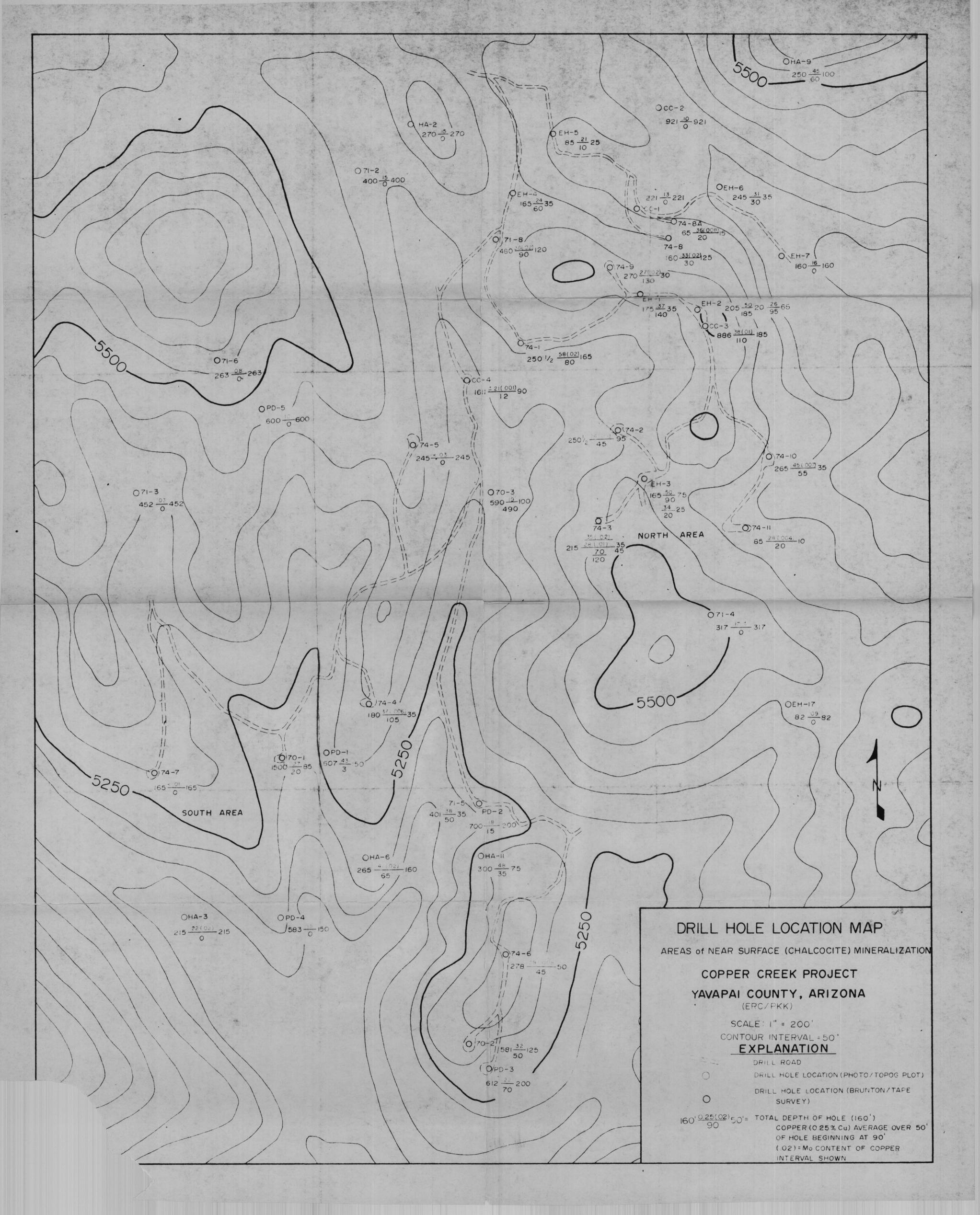
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		V.V	107758 LD 207640*LD	ASHBY I	SZCZOTKA THOMAS IRWIN CHARLES	207632	1572:664	9/01/1983	1886	2/07/1986
	NW 25	٦	207641*LD	٩	IRWIN CHARLES	207632		9/01/1983	1988	
			1	LAME #18	JONES DAVID	248283	1739:380	1/20/1986		5/25/198
	NE 25		248302*LD 248302*LD 249024*LD	LIGHTNING FLAME #19 LIGHTNING FLAME #20 LIGHTNING FLAME #25		248283 248283 248283	1789:382 1789:384 1796:79	1/20/1986	1-0 0 90 0 80 0 80 0 80	5/25/1988 12/26/1989
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	20 NE 25	5	26581 PL	YOUNG MINING CO #1	YOUNG MARVIN E		1151:340	7/10/1978	1979	6/06/1983
	N2 25	5	26582 PL	YOUNG MINING CO #2	COLLIE, JERRY YOUNG, MARVIN E	26581	1151:339	7/10/1978	1979	6/06/1983
		.,	9	2	COLLIE, JERRY		- 1		- 1	
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Opper Creek

Deal

Now perging

15 plat + name ambent

Two outstand parts

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Repile Chalcacite Chalcapyrite Cornellite

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20mm in two reprarated backers
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Then 2200/mo "177

4125/Idom team

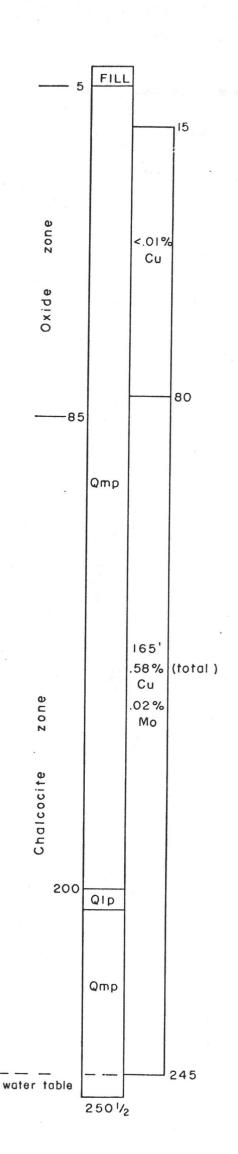
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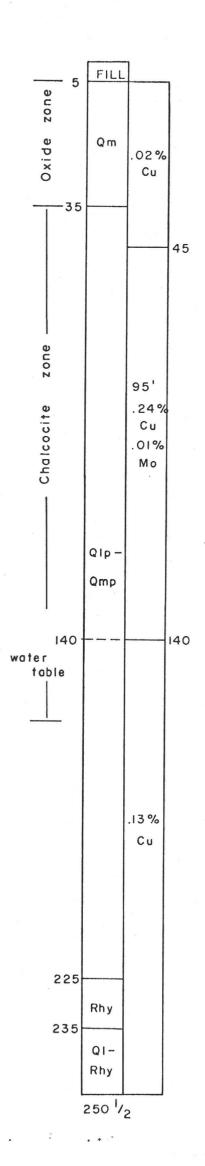
50,000 remains.

Contre Peramer Sive a one year orphin which guarantes payment of all abligations If went to centime begins 192 pay \$100,000 for two more gene at end of third year pay 500,000 mare Everyone out. Buyout frie from carter - 1600,000



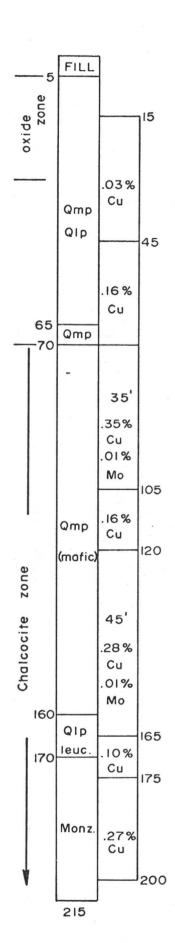
Essentially quartz monzonite porphyry; siliceous with numerous quartz veinlets; moderate to heavy chalcocite 90-200. Very minor chalcopyrite 145 to bottom. Good secondary biotite above 200. Mosa stringers 160-200. Pyrite variable but generally +1%.

This is the best hole of the CC 74 series holes.



Quartz monzonite porphyry-quartz latite porphyry; heavily silicified; some chloritization; better than average LoS₂ (especially near 220-235 dike contacts).

1-3% pyrite; trace to some chalcopyrite.

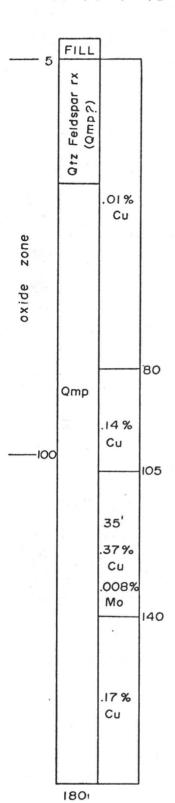


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Summary:

Mafic Quartz latite porphyry to quartz monzonite porphyry to 160. Monzonite (low in pyrite) 160 to bottom.

Trace to minor chalcocite 35-215. 1-2% pyrite, trace chalcopyrite.

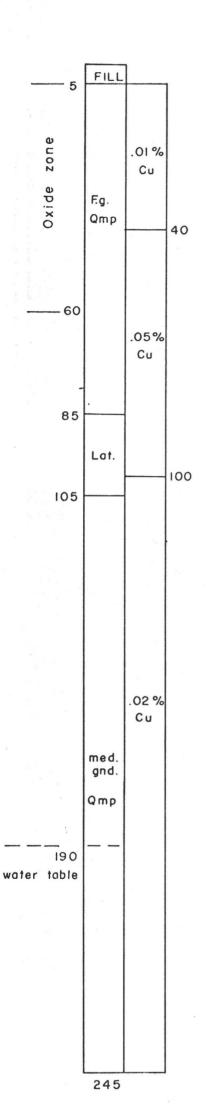


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Summary:

Entire hole is probably quartz monzonite porphyry. Good silicification. K-feldspar flooding 160-170.

2-3% pyrite, diminishing below 115'.
MoS₂ in quartz veinlets 150-155'.



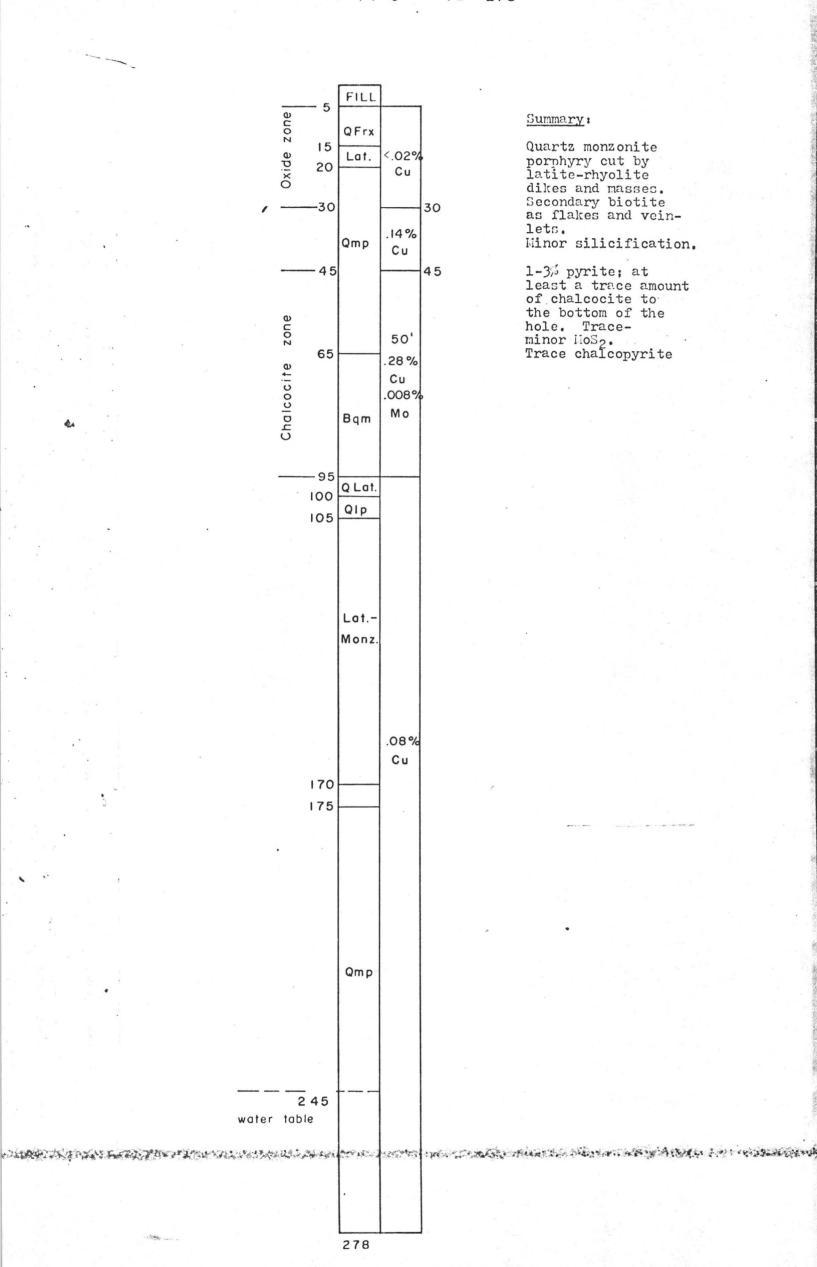
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Summary:

Hole located outside well-mineralized area. Some propylitization.

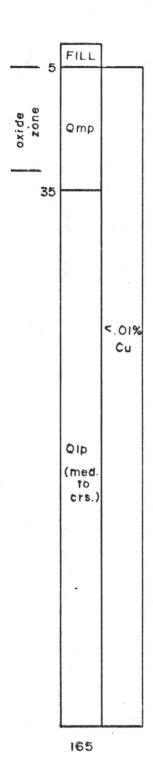
1/2-2/3 pyrite, bare trace chalcocite 25'-130'; trace chalcopyrite 45'-60', more below 190'. Scattered traces MoS₂.

A duster!



Quartz monzonite porphyry cut by latite-rhyolite dikes and masses. Secondary biotite as flakes and veinlets. Minor silicification.

1-3% pyrite; at least a trace amount of chalcocite to the bottom of the hole. Trace-minor MoS2. Trace chalcopyrite



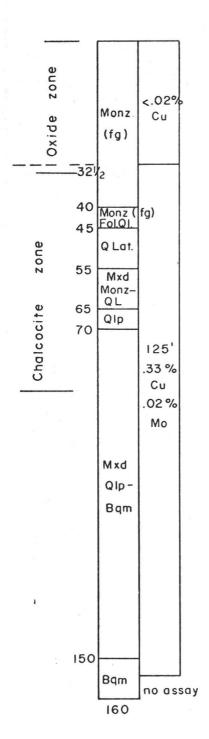
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Summary:

Quartz monzonite porphyry-quartz latite porphyry.

Only trace amounts of pyrite encountered. Water table not encountered.

A duster!



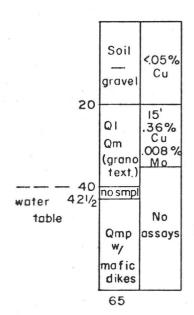
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Summary:

Mixed rock types; well-broken (faulted) Propylitized - with possible secondary biotite

+3% pyrite (heavy for area)
Heavier than normal chalcopyrite

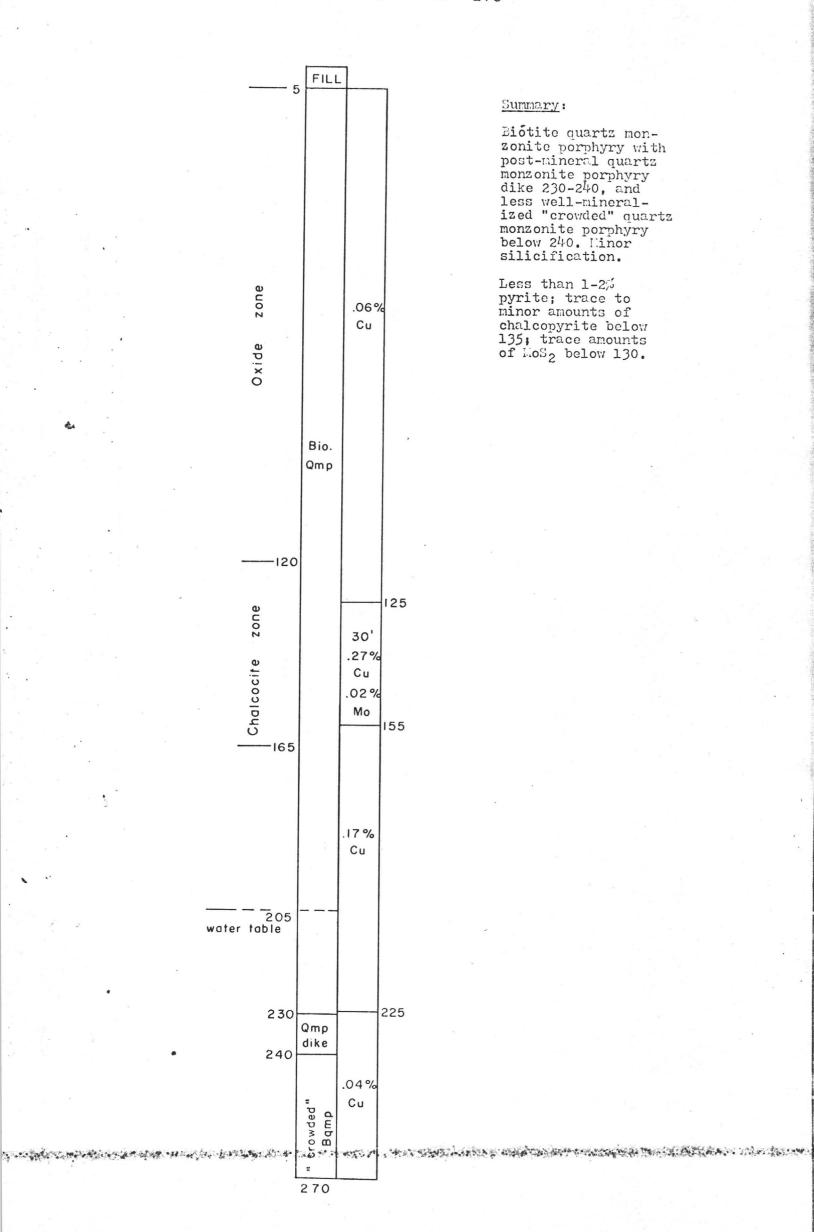
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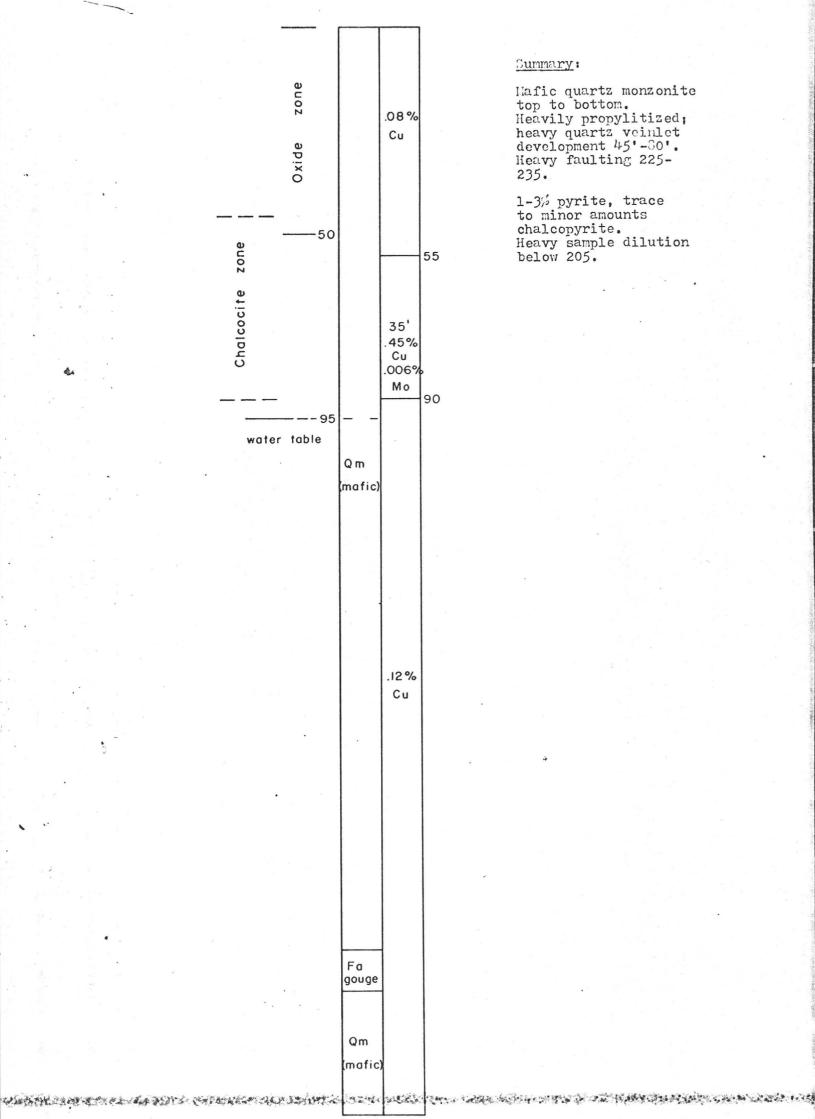
Mixed rocks; well fractured.
Some propylitization

2-5% pyrite (heavy); locally good chalcocite, some chalcopyrite and MoS₂



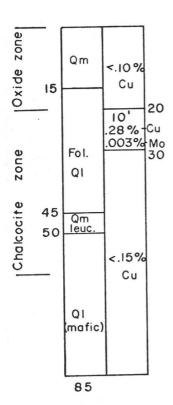
Biotite quartz monzonite porphyry with post-mineral quartz monzonite porphyry dike 230-240, and less well-mineral-ized "crowded" quartz monzonite porphyry below 240. Minor silicification.

Less than 1-2% pyrite; trace to minor amounts of chalcopyrite below 135; trace amounts of MoS₂ below 130.



Hafic quartz monzonite top to bottom.
Heavily propylitized;
heavy quartz veinlet development 45'-80'.
Heavy faulting 225-235.

1-3% pyrite, trace to minor amounts chalcopyrite. Heavy sample dilution below 205.



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Summary:

Well-chloritized quartz latite cut by quartz monzonite dikes. Some quartz veinlet development.

2-3% pyrite; trace chalcopyrite and MoS₂.