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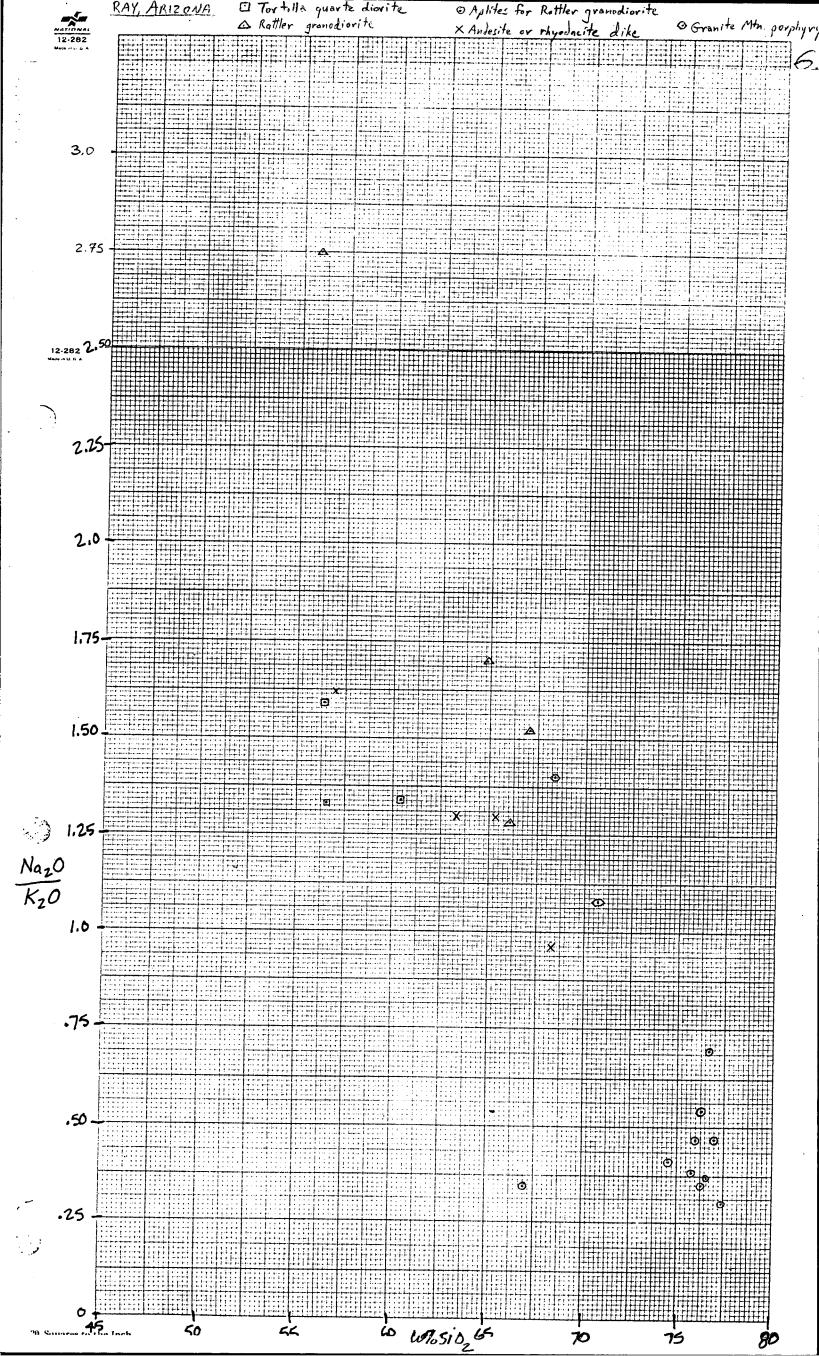
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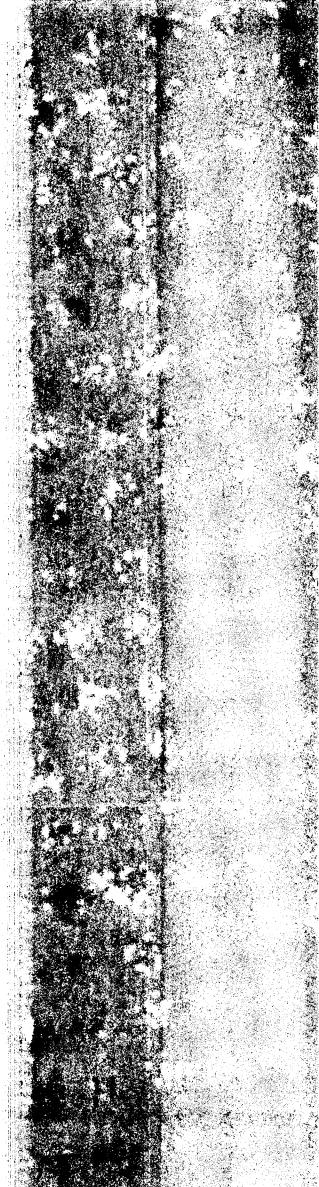
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MINE NAME - MARY PIPER CLAIMS MINE GPERATORS - GRADY L.HERRING 22132 ORE TREATED (SHCRT TONS) (SHORT TONS) COUNTY: PINAL (SHOPT TONS) (SHORT TONS) 120. 14. 16. • • • |--|--| COUNTY: PINAL MINE NAME - LEE WONG MINE OPERATOPS - J.RICHARDS MINE NAME - LITTLE TREASURE MINE OPERATORS - N.H.MELLER MINE NAME - LAWRENCE MINE OPERATORS - A.B.COOK l 1 ł YEAR TYPE 1935 NC TOTAL S YEAR TYPE 1922 NS 1923 NS 1924 N'S 101AL S YEAR TYPE 1931 NG 1959 NG 1963 NG 107ALS YEAR TYPE 1917 NS 1921 NS 1921 NS 1922 NS 107ALS ŧ € $\left(\right)$ ¢ M D E Z S Complete Field was a was a total available. 6) ŧ ţ ŧ ţ ¢ Ç Ą

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MINE NAME - SENATOR GRP. AB E FERA (+ C. FERA, 1041) A. TRAFANOVICH, LEASEE, 1929 MINE CRERATORS - AB E FERA (+ C. FERA, 1041) A. TRAFANOVICH, LEASEE, 1929 YEAR TOPE (SHBRTTADIS) (ODULDES) (ODULLASS) (DULLASS) YALY YEAR TOPE (SHBRTTADIS) (ODULDES) (ODULLASS) (DULLASS) YALY YEAR TOPE (SHBRTTADIS) (ODULCASS) (DULLASS) (DULLASS) (TOUNCESS) YEAR TOPE 131. (STOUL) (SULLASS) (DULLASS) (SULLASS) YALY YALY 1922 131. (STOULASS) (DULLASS) (SULLASS) (SULLASS) (SULLASS) YALY 1942 1139 1139 1139 1139 1129 1230 YALY 1942 YALY 1479 1169 90 123 123 YALY YALY YALY YALY YALY YALY YALY YALY YALY YALY YALY YALY YALY YALY YALY YALY YALY YALY YALY YALY YALY YALY YALY	MINE CPERATORS - TAME - SENATORS GRP. MINE CPERATORS - TOR GRP. MINE CPERATORS - TORS GRP. MINE CPERATORS - TORS GRP. VEAR T GPE VEAR T GPE (900LLARS) (900LLARS) 19420 19442 19442 19442 19442 19442 19442 19442 19442 19442 19442 19442 19442 19442 19442 19442 19442 19442 19442 19442 1944 <td< td=""><td></td><td></td><td>06350</td><td>571213</td><td>7952</td><td>6436</td><td>18</td><td>15</td><td>339</td><td>0404</td></td<>			06350	571213	7952	6436	18	15	339	0404
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	ת\ •		SRE TREATED	CHRISTMAS M CO		201. 201.		OUEENS FRANK HUBBELL	0	10	ORE TREATED	ER THOM R.S. FISHER	U 1	15	CRE TREATED	ER CROSS PEDERSON+GARLAN	COUNTY: PINAL		
	17.			** * * * * * * * * * * *	0	EC 23 E C 23 E C 34	(POUNDS)	CENTR&L *+0			(ODDEP)	+++ + + + + + + + + + + + + +	•		(POUNDS)	GNINIM + D	75		
		IN	(DOLLARS)	+++++++++++++++++++++++++++++++++++++++	(n	6 00	1 vim	• + + + + + + + + + + + + + + + + + + +	о •		(DGLLARS)	+++++++++++++	0	0	(DOLLARS)		INING DISTRICT		
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	01	60	(DOLLARS)	* * * * * * * * * * * * * *	179.		Sm	++ ++ ++ ++ ++ ++ ++		39	SIL VER	· + + + + + + + + + + + + + + + + + + +	сл I Со I	391	(DOLLARS)				
		0	(3 L)	. + ++ + ++ ++ + + + + + + + + + + + +	- 01	1	5	* * * * * * * * * * * * * * * *	0•	0	(DUNCES)	* + + * + * + * + * + * + * + *			(DUNCES)				
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(13				د روم محمد				
MINE NAME - UNKNOWN	JNKNDWN 25 - P MUNGREY			L r				
YEAF T	URE TREATED (Short Tons)	(SUNDER)	(CCPPER)	SILVEP (JUNCES)	SILVER (DJLLAFS)	(JUNCES)	(SALULARS)	YEARLY VALUE ALL METALS
1920 NS		50	0 1 1 1	0.0	363		21	32
TOTALS				00	0.0		21	12
++++++++++++++++++++++++++++++++++++++	++++++++++++++++++++++++++++++++++++++	+ + + + + + + + + + + + + + + + + + + +	++ ++ ++ ++ ++ ++	+ + + + + + + + + + + + + + + + + + + +	* + + + + + + + + + + + + + + + + + + +	• + + + + + + + + + + + + + + + + + + +	++++++++++	+++++++++++++++++++++++++++++++++++++++
т С	OPE TREATE (SHORT TONS		CCPPER (CCLLARS)		illo.	(300 C)	(DJLLARS)	
1927 N		52	m 	191	30			0.66
	0	5		[m]	30	• 0		080
-	ORE TREATED (SHORT TONS)	L E A UND S	(COLLARS)	(POUNDS)	(COLLARS)	(SONDCo) Word on Con	MJLYBJENUM (DJLLARS)	YEARLY VALUE ALL METALS
1927 N	50	14020	276					086
1 _ 1	10	4020	171					0
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e T YP	ORE TREATE (SHORT TONS	と と し	11102	n.s	NE	(JUNCES)	(DOLLARS)	AL U TA L
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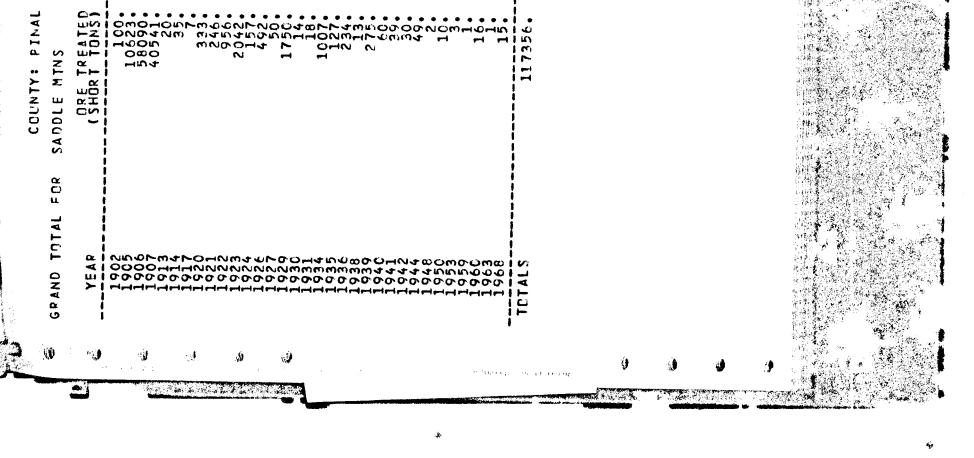
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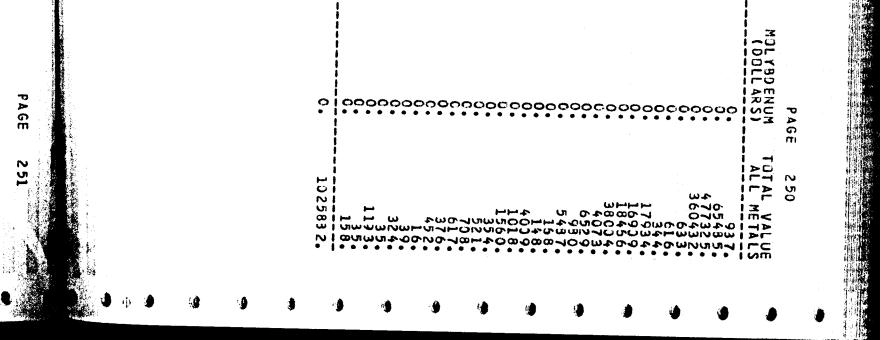
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IGNEOUS INTRUSIONS AND ASSOCIATED MINERALIZATION IN THE SADDLE MOUNTAIN MINING DISTRICT PINAL COUNTY, ARIZONA

by

Larry Frank Barrett

A thesis submitted to the faculty of the University of Utah in partial fulfillment of the requirements for the degree of

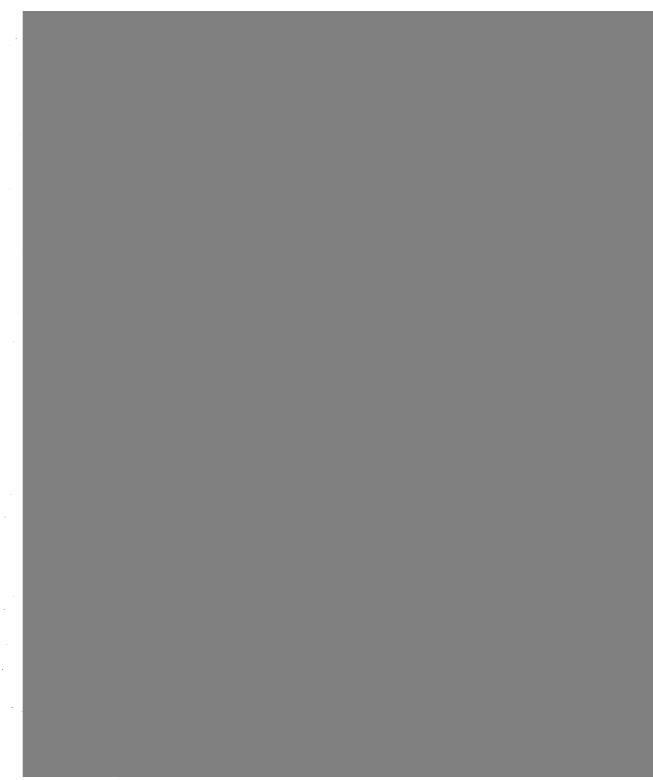
Master of Science

Department of Geological and Geophysical Sciences

University of Utah

June 1972

ACKNOWLEDGEMENTS



WILLIAMSON CANYON VOLCANICS

DATA SHEET FOR CHEMICAL AND AGE DATA

FOR IGNEOUS COMPLEXES

LOCATION:

Williamson Canyon Volcanics, SE Arizona PINFL COUNTY

LOCATION COORDINATES:

KG: - CA-AC [most in CA] fields. $K_{55} : 2.65$ 2.8 $K_{57.5} : 3.15$ - CA-AC fields K Mg: ~ 57.0 AC Fe: weakly nch A1: 0.665 - 0.950

PEACOCK INDEX:

Metaluminous, alkali-calcic or calc-alkalic, weakly iron-rich.

ROCKS INVOLVED:

AGE DATA: 80-74 M.Y. [TOMBSTONE ASSEMELAGE - HOjave Metaluminous] COMMENTS: Rocks related to Williamson Canyon volcanics have three dates ranging from 80-

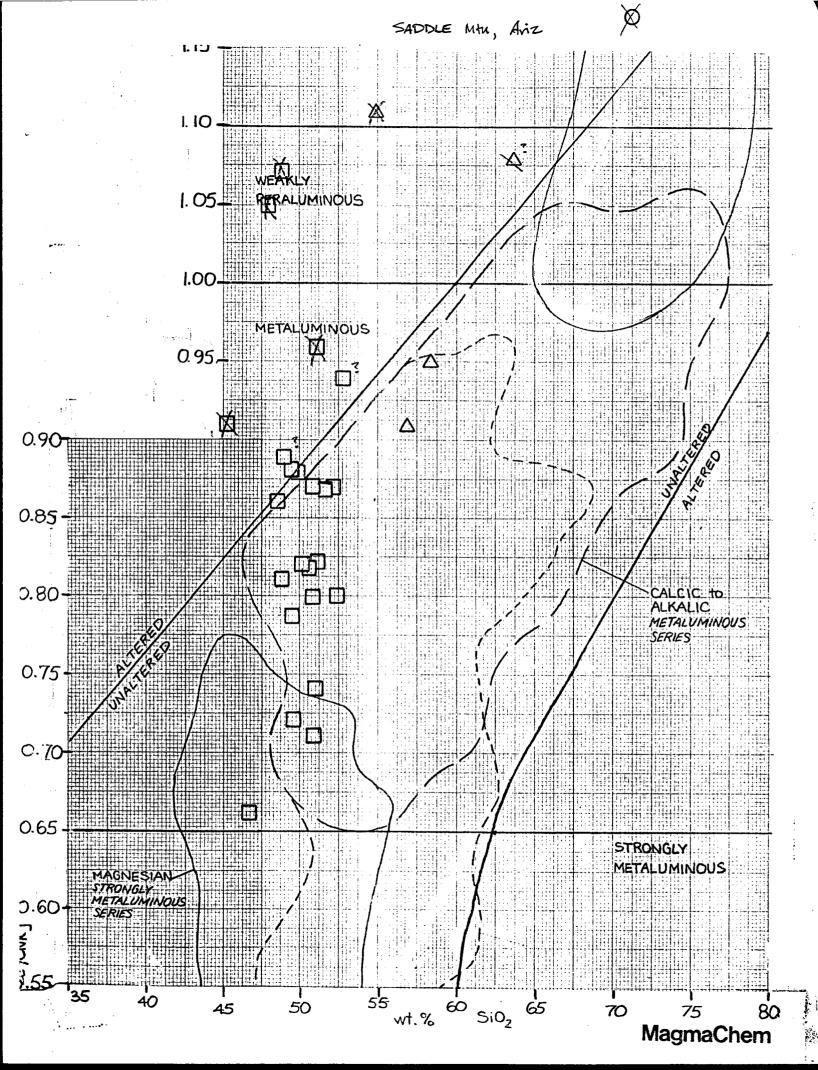
volcanics have three dates ranging from 80-74 m.y. Volcanics are intruded by probable 72-70 m.y. calcalkaline intrusions.

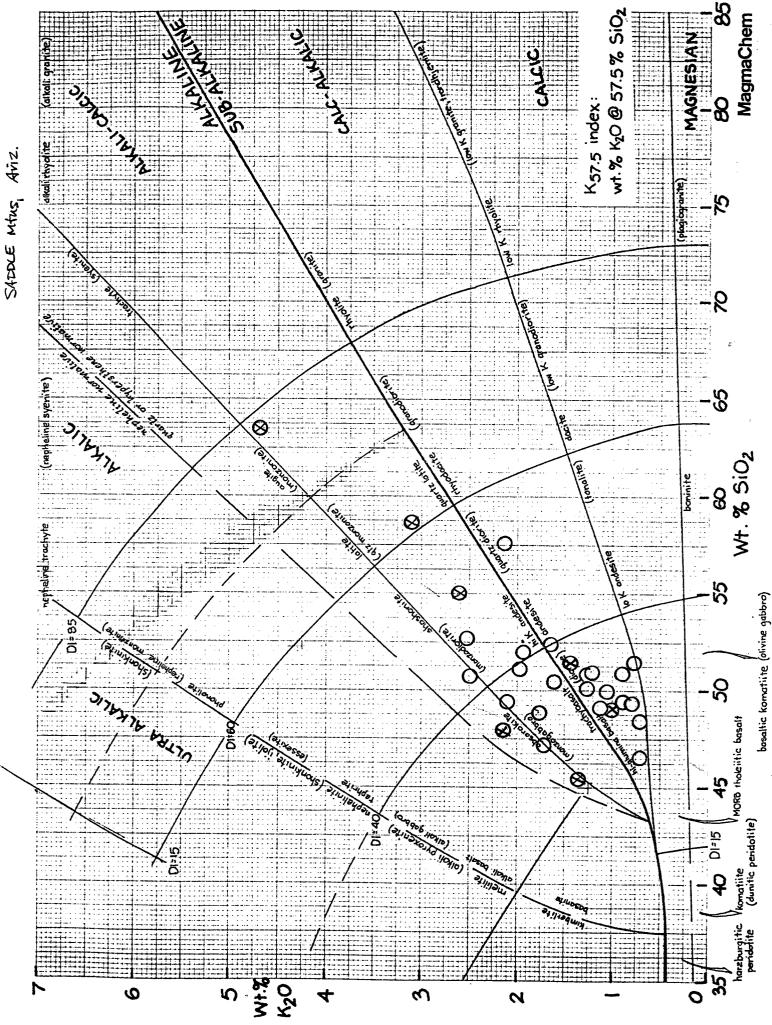
REFERENCES FOR CHEMICAL DATA:

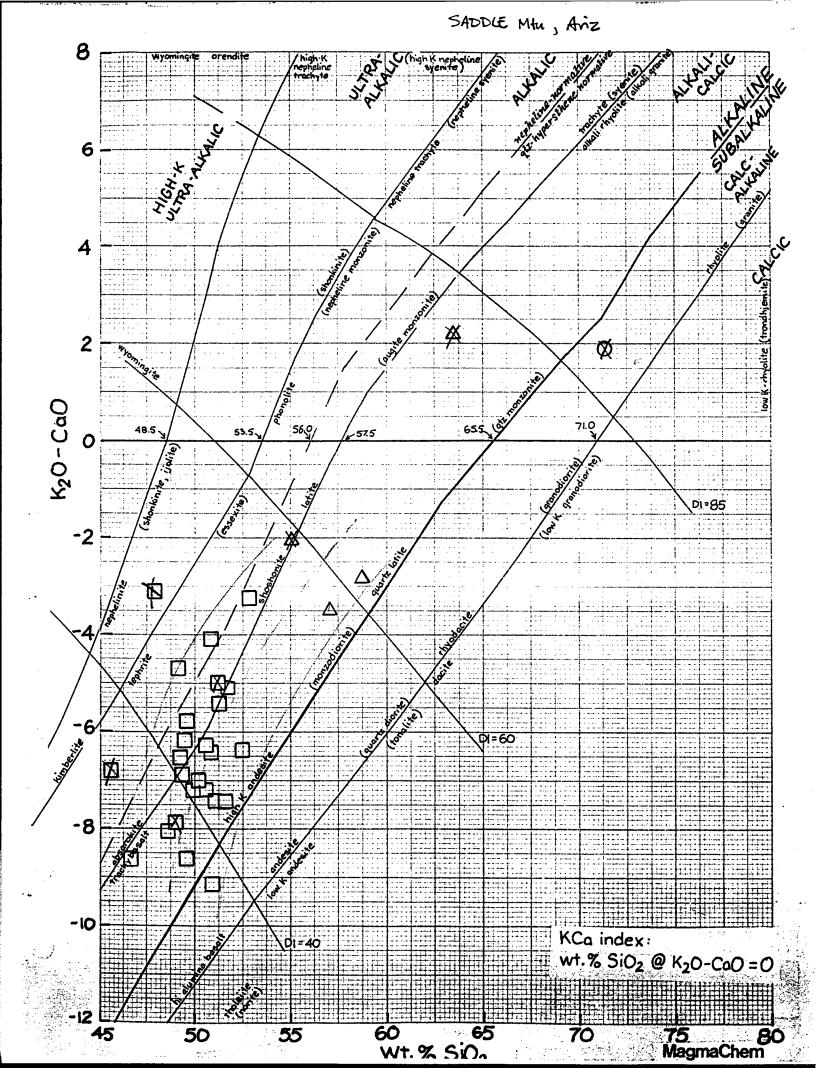
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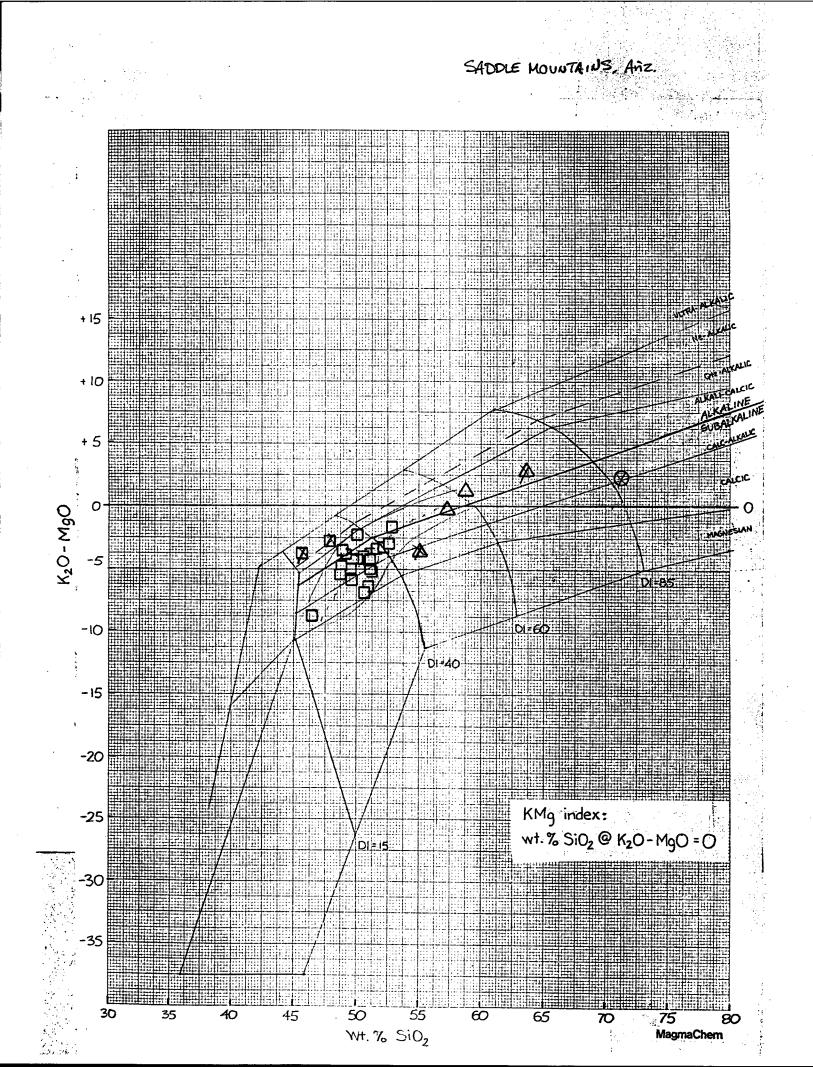
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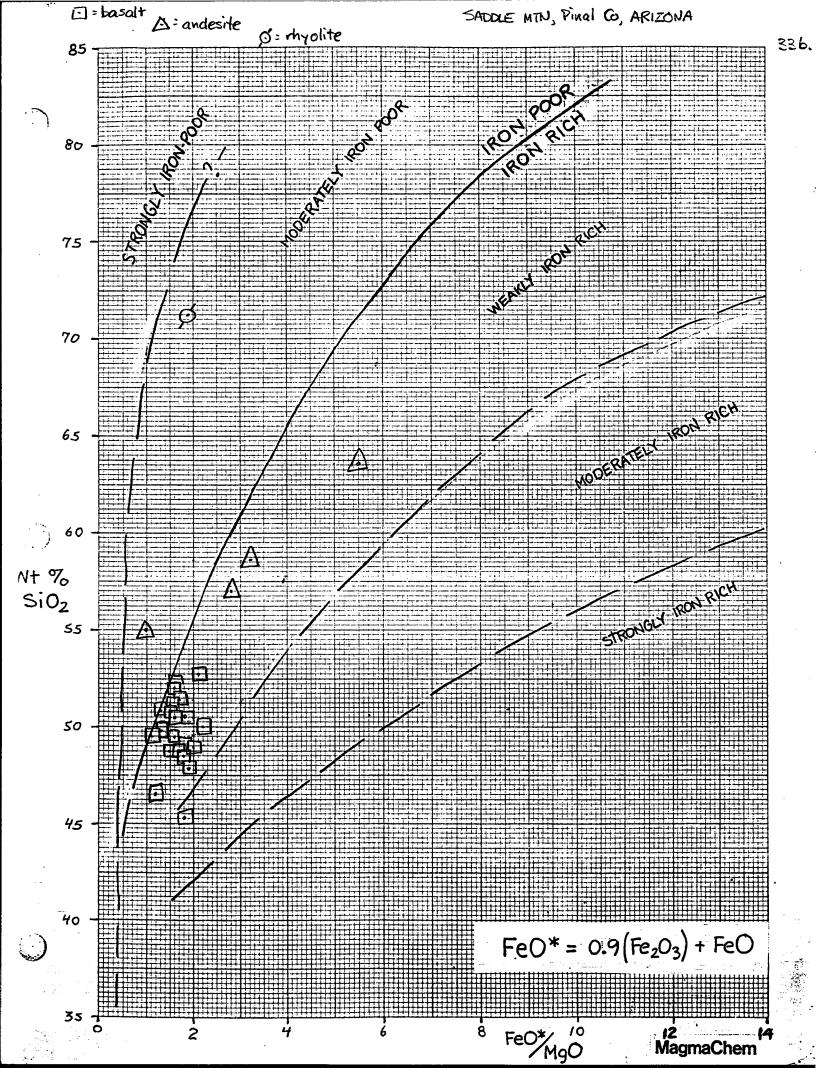
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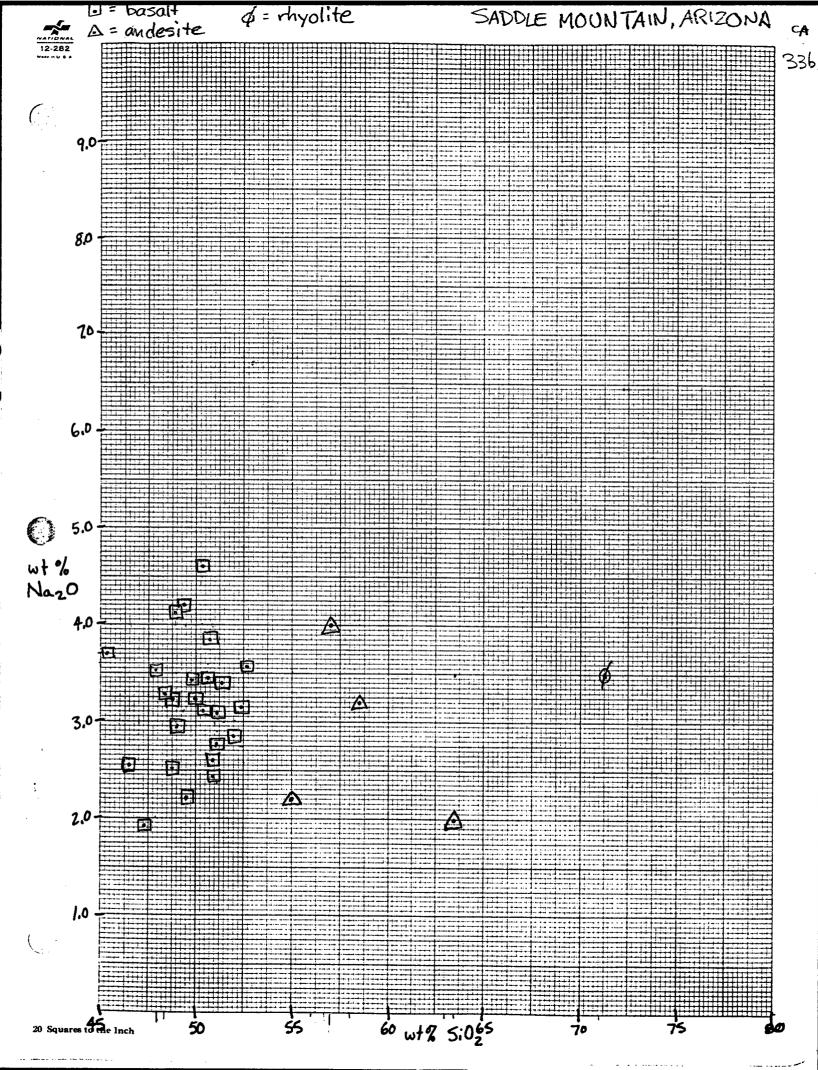


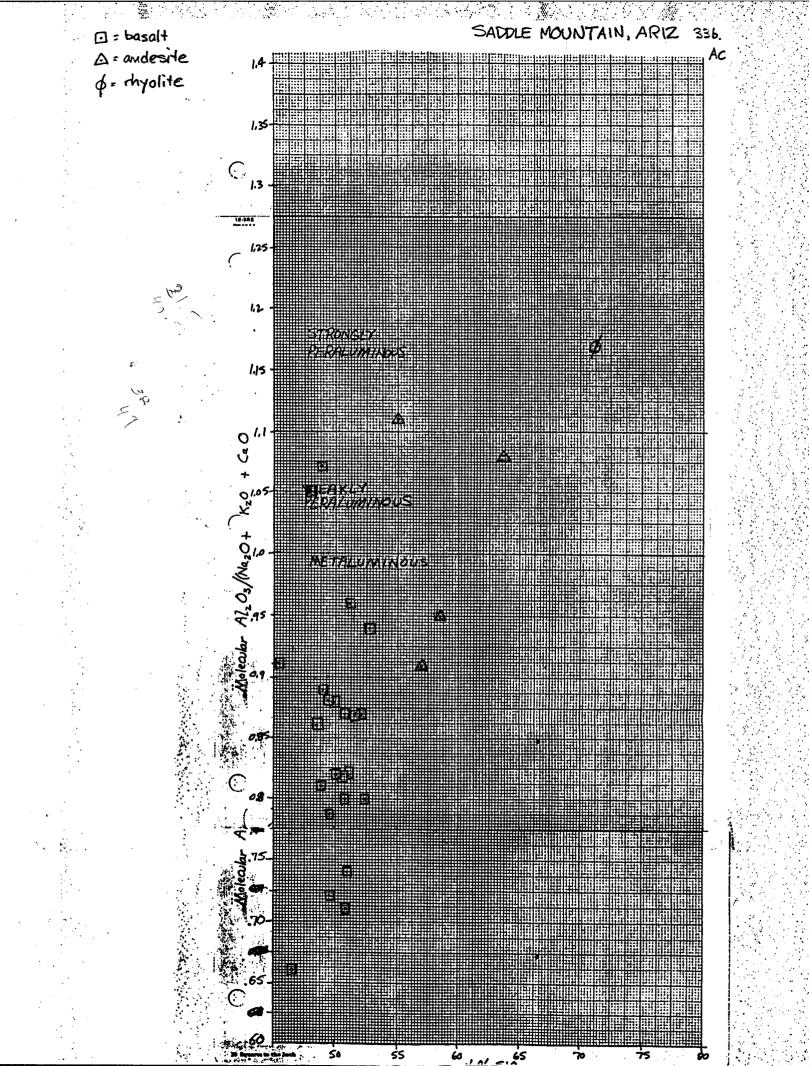


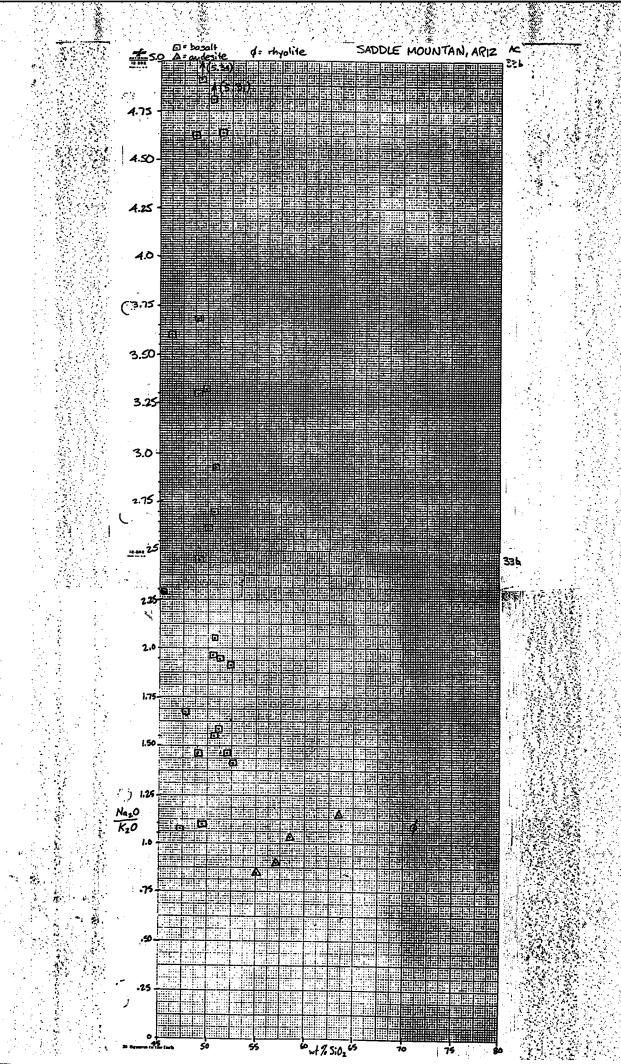




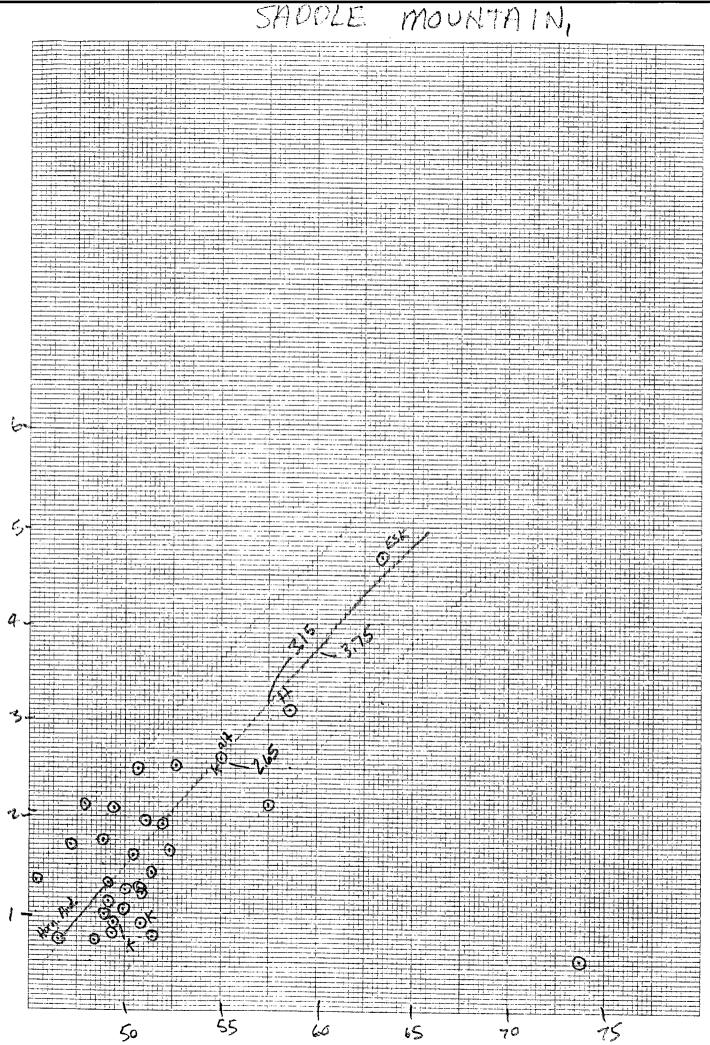








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Kamecoli Exploration, Inc.

Exploration Services Department

Geochemical Research and Laboratory Division

March 11, 1975

L. Clarger

MEMO TO: John C. Wilson

FROM: Lloyd A. Clark

SUBJECT: Chemical Classification of the Williamson Canyon Volcanics, Pinal Co., Arizona by Larry B. Clayton

In this phase of the 1974 Andesite Research, it was hoped to define the eruptive center of the volcanic edifice, and to establish relations between volcanic differentiation, genesis of the Laramide stocks and dikes, and the O'Carroll Canyon sulfide system. However, the several thousands of feet of volcanic rocks are all basalts and are undifferentiated. Probably the original volcanic pile was several times thicker and more laterally extensive. It now is apparent that erosion has removed that part of the volcanic pile necessary to accomplish the research objectives.

Lloyd A. Clark

LAC/db Attachment

cc: R. C. Babcock

N. A. Gambell

R. L. Nielsen

G. D. Van Voorhis

CHEMICAL CLASSIFICATION OF THE WILLIAMSON CANYON VOLCANICS, PINAL CO., ARIZONA

by

L. B. Clayton

February, 1975



SUMMARY

A six-mile traverse was completed starting at the Gila River and continuing east along the border between T. 4 S. and T. 5 S. into Reed Basin (R. 16 E. and R. 17 E.). Fresh rock chip samples were collected approximately every fifty feet in composite samples over intervals of 500 to 2,000 feet. The traverse crossed the entire north limb of the Deer Creek Syncline and continued down the plunge of the Syncline to the east. This represents a volcanic section in excess of 3,000 feet. At least 85% of the section sampled was pyroclastic in origin. Fragment size varied from lapilli to breccia blocks, usually subangular. Only two flow units were observed in the pile.

Whole rock chemistry, heavy liquid separations, X-ray diffraction and thin section studies indicate that the volcanic pile was derived from a subalkaline magma. The rocks can further be classified as calc-alkaline based on high alumina content and location on an AFM diagram. Plagioclase, the common phenocryst mineral, also supports the calc-alkaline classification. However, the common lithologic variation normally observed in calc-alkaline volcanics (basalt-andesite-dacite-rhyolite) is not developed. All samples collected are basalts, which suggest that the majority of the volcanic pile has been eroded away.

INTRODUCTION

The Williamson Canyon volcanic pile was studied to determine if the source volcano could be located and genetically related to the O'Carroll sulfide system or any other system, or if the volcanic pile is early Cretaceous in age simply acting as a host to mineralization.

The normal volcanic differentiation sequence of basic to intermediate to felsic flows and pyroclastic is an easily mappable feature which can be verified geochemically. From these data an eruptive center can be predicted. The volcanic differentiation is usually followed by emplacement of stocks and plugs in and around the eruptive center, and if located, may provide areas with high mineral exploration potential. The late stage magmatic activity is often expressed as a dike swarm along a major structural trend similar to that at O'Carroll Canyon.

A total of nineteen rock chip composite samples were collected. These are combined with other data on volcanic rocks available from drill hole OC-1 and Hansen ddh (Fig. 2). The samples represent a volcanic section in excess of 3,000 feet across the north limb to the axis and down the plunge of the Deer Creek Syncline (Fig. 1). Every attempt was made to omit obviously mineralized material from the samples. Samples were collected as far as possible from dikes and mineralized fissures. Approximately one fragment (3-5 cm) of rock was collected every 50 feet after chipping away the weathered surface. Two to four pounds of sample were collected for each composite (Fig. 2).

Whole rock wet chemical analyses, spectrographic (semi-quantitative) analyses (for possibly anomalous metals), thin sections and X-ray diffraction work were performed on the samples (see Appendix).

DISCUSSION

Field observation indicated that 85% of the rocks collected are pyroclastic in origin. Fragments range in size from lapilli to breccia blocks usually subangular. Only two flow units were observed in the pile. One occurs at the base of the pile (Sample 1) and is a dense, black aphanitic unit, several hundred feet thick. The second flow unit is amygdaloidal, also several hundred feet thick (Samples 6 & 7), and lays conformably upon interbedded sediments within the volcanic pile at the head of Little Gold Gulch. Mineralization was observed in the pyroclastic rocks above the amygdaloidal flow (Samples 8, 9, and 10).

Chassification based on chemical composition indicates that the balk of the volcanic rocks falls in the subalkaline field (Figure 3a) and can be further classified as calc-alkaline based on the AFM diagram ($A=K_2O+Na_2O$, F=FeO ± 0.8998 Fe₂O₃, M=MgO) (Fig. 4a). The calc-alkaline classification is further supported by the high alumina content which is the most prominent chemical difference between more basic volcanic rocks. "The calc-alkaline basalts and andesites are generally high alumina types containing 16 to 20% Al₂O₃, whereas their tholeiitic counterparts have only 12-16%" (Irvine and Baragar, 1971). The average for the Williamson Canyon pyroclastics is 18.13% Al₂O₃. Further classification, based on silica content (average pyroclastic 50.55% SiO₂) and CaO - K₂O - Na₂O ratio (Fig. 4b), indicate that the rocks are basalts. The weakly mineralized samples at the head of Little Gold Gulch (Samples 8, 9, and 10) show increases in SiO₂, K₂O, and possibly Na₂O relative to the unaltered samples.

Brief examination of thin sections in most cases confirms the pyroclastic texture of the samples. However, some samples that were obviously pyroclastic in outcrop appear to be flows in section because of large fragment size. Fresh plagioclase is present in Samples 1, 6, 12, and 13. In the remaining sections plagioclase shows varying degrees of clay alteration, presumably as a result of weathering or possible low grade metamorphism. Chlorite occurs in most samples as a result of alteration or weathering the mafic minerals. Little quartz was detected in thin section; however, X-ray diffraction indicates the presence of quartz in most samples.

Heavy liquid separations (> 2.8 gm/cc) were made and X-rayed on eight of the nineteen samples (see Appendix). Results indicate the presence of chlorite in most of the samples. Magnetite was detected in all the samples and pyrite was present in Sample 18 along with trace amounts of epidote, chlorite, and hematite. Pyroxene (augite) is present in most of the samples taken at and above the amygdoloidal flow with amphiboles present throughout most of the samples (both X-ray and thin section). Olivine was not detected in any of the X-ray or thin section work.

Reference

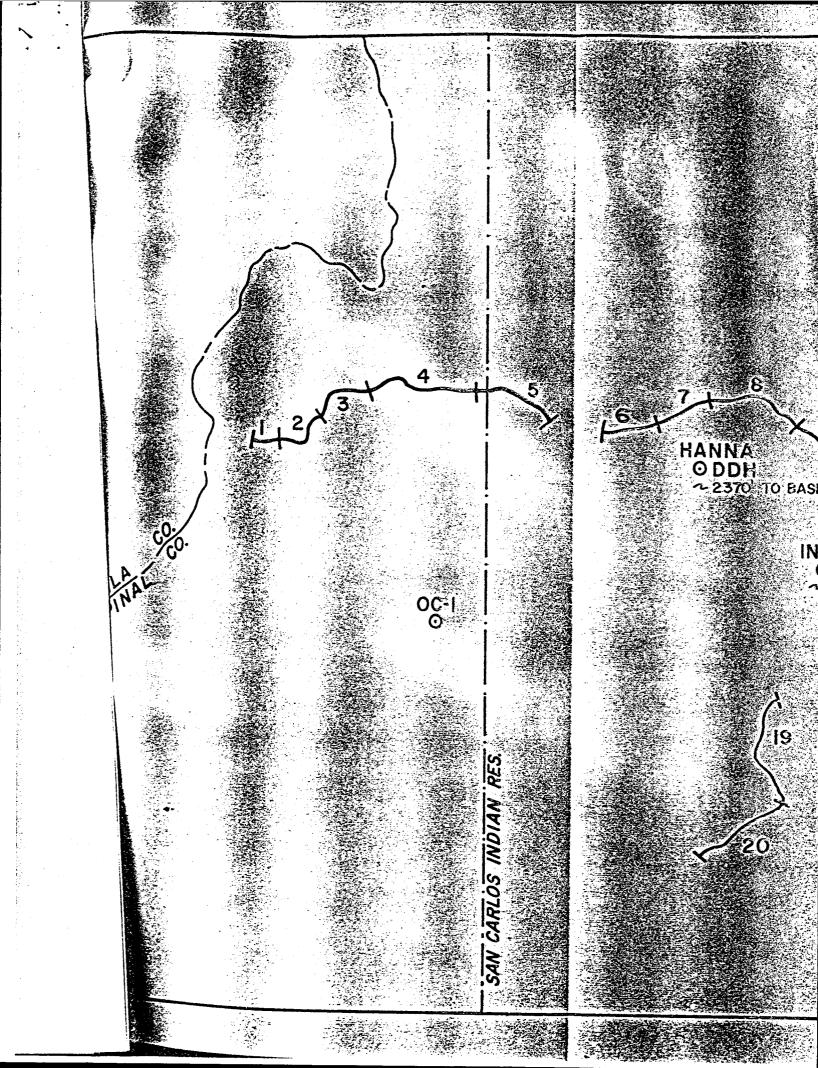
Irvine, T.N. and Baragar, W.R.A., 1971, A guide to chemical classification of common volcanic rocks: Canadian Jour. Earth Science, v. 8, p. 523-548.

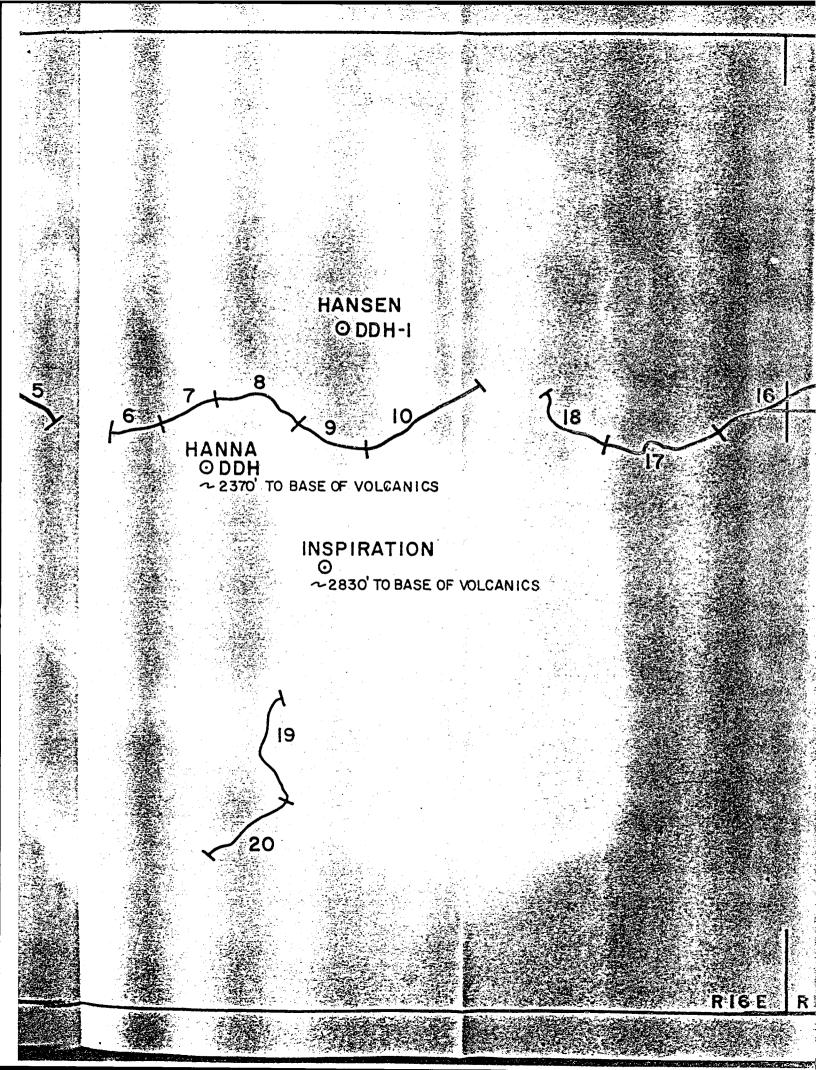
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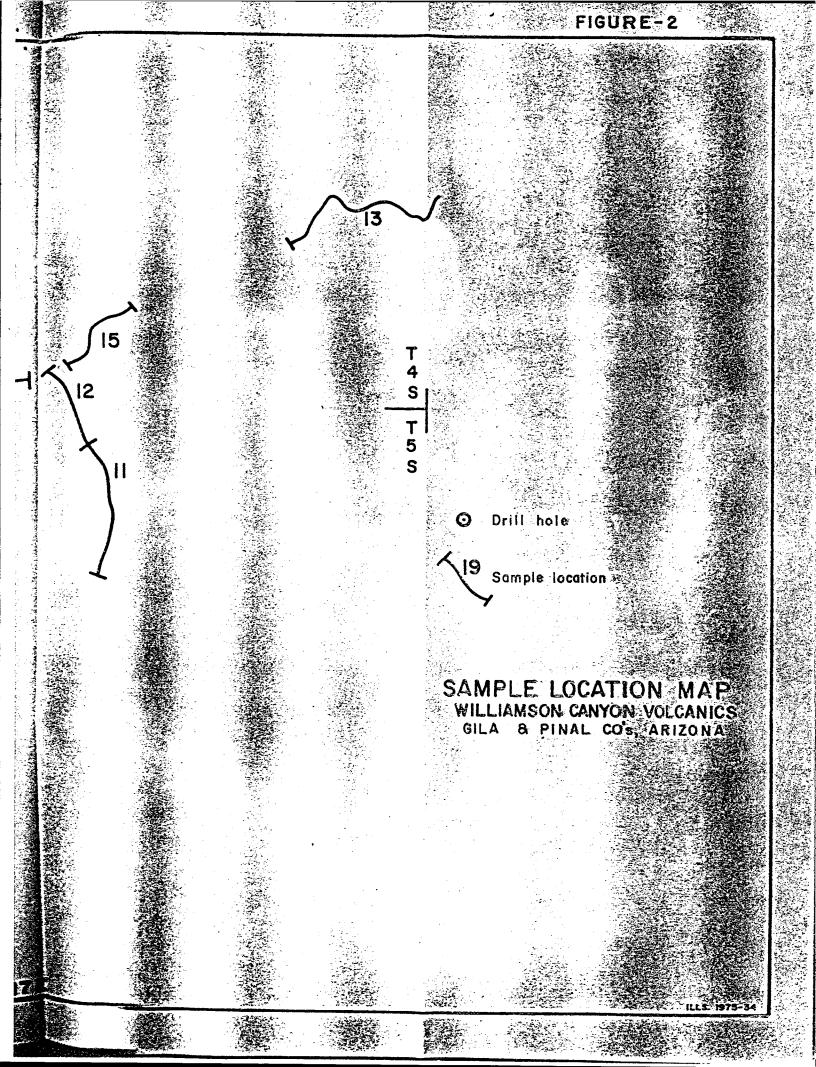
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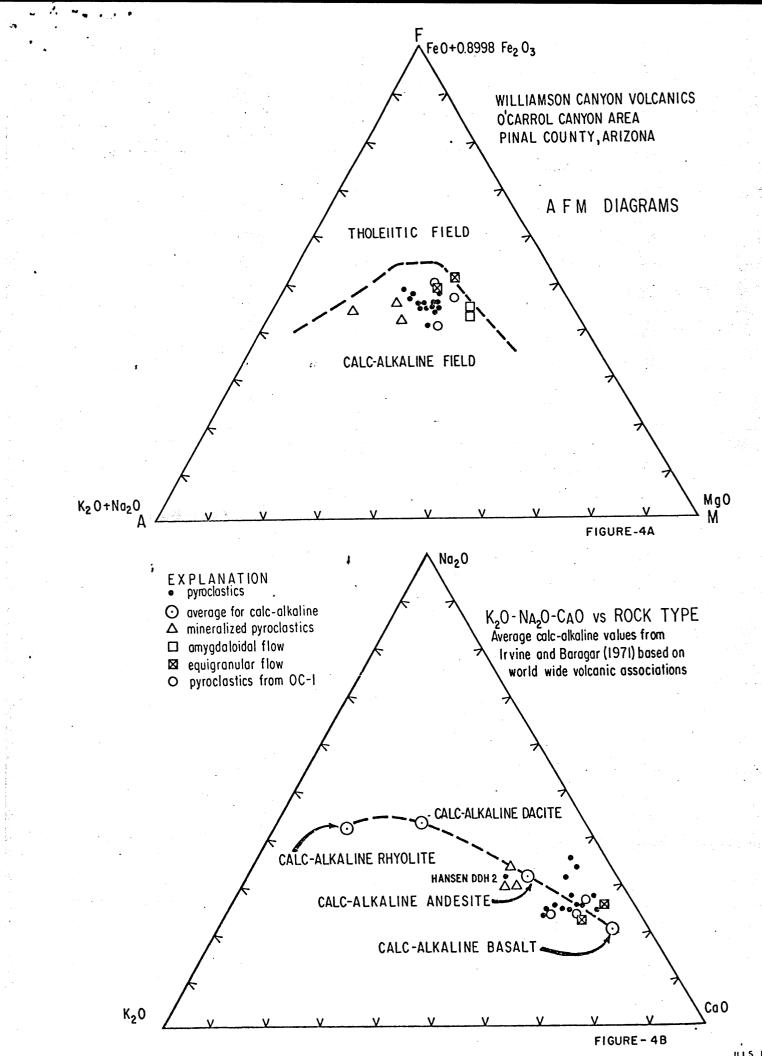
APPENDIX - Wet Chemical, Spectrographic and X-ray Analysis of Williamson Canyon Volcanics

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				٤ ⁰ 5 ¹	v 9	17.6	19.4	18.9	19.1	18.1	15.7	14.7	18.5	18.2											16.9	18.9	19,3	19.0	19.3	19.7
•.			÷	^z o:	s :	* ° *	50.4	\$1.4	43.9	19.3	50.9	49.4	50.7		57.0			, .	0.0	48.8	-	1.10 1.10		1.10	50.4	47.9	48.8	49.1	49.8	45.3
		-				0CA- 1	OCA- 2	OCA- 3	OCA- 4	OCA- 5	OCA- 6	0CA- 7	004-8	- 400				000	0CA-13	004-16				41-17	OCA-20	T-4592 Hanson DDH	Γ Τ-5527	oc-1-81715829	36. Poc-1-865: 2893.	31. loc-1-1101120.
				5	111		44.1	36.45	4507	44.5	33,35	36	505	40.95	£ 8	41.15	40	40.9	41.35	37.75	41.5	39,8	40.5	49	<u></u>	\[344	1-20 gregte	36. foc-1	31, 1 ₀₆₋₁

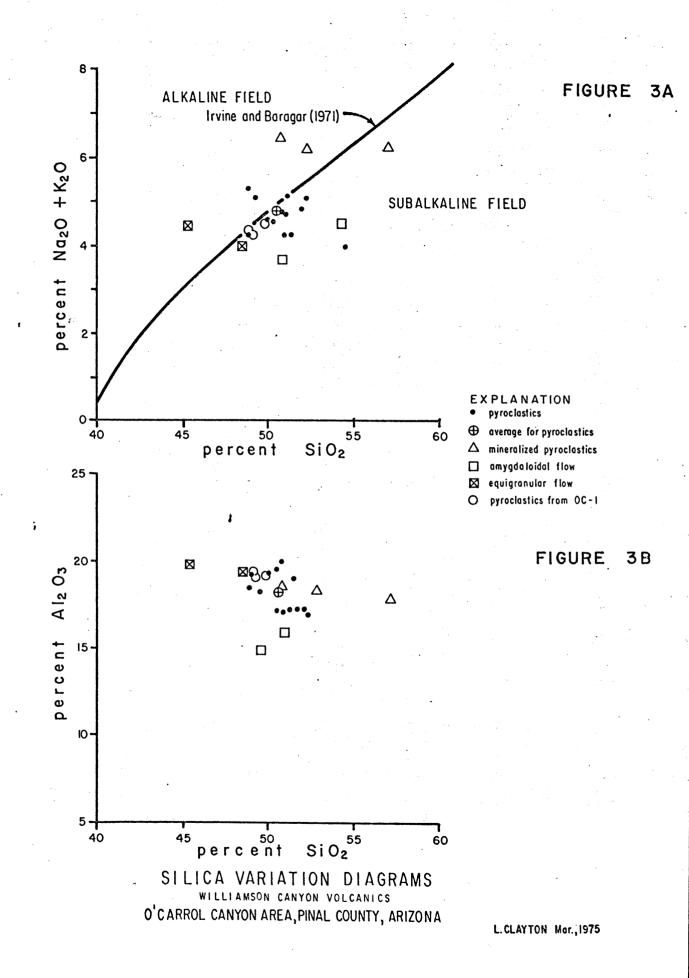








ILLS 1975-36



ILLS. 1975-35

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Plag., Qtz., Mte., Augite	800	475	. 08	<u>^</u>	.7	:		و :	78	:	=	:	=	:	5	90	69	2 ~ . 2	4, 02	52	3.11 1.59	7.87 3.	5.58 7.	2.02 5.	36	16.9 7.	50.4	OCA-20
NA	700	575	. 085	. 17	.7	:		9 -	98	• •	=	=	:	:	7 .	96	86	2.48	3. 02	5 102	3.09 1.95	7.39 3.	5.71 7.	29	.96 1.	17.1 7.	51.1	OCA-19
Plag., Qtz., Amphibole, Chl., Mte., Hematite, Pyrite?, Epidote		350	. 095	=	.6			• •	34	:	-	с. С			6 :	132	70	4.76	5 2.44	7 1665	3.31 1.27	7.99 3.	4.83 7.	98	.07 2.	19.9 4.	50.8	OCA-18
NA		425	. 095	:	.7		-	11 °	60	:	=	:	: '		œ •	89	85	6 2.76	4.56	2 71	77 1.42	6.38 2.	4.91 6.	1.85 4	6.42 1.	17.3 6	51.1	OCA-17
NA		280	. 09	:	. 7	:		10	30 . 10	.=	÷	:	 3	:	ۍ -	80	96	8 2.35	2 3.58	32 52	24 .982	8.84 3.	30	2.16 4.	60	18.4 5.	43.8	OCA-16
Plag., Mte., Augite, Chil?,		490	. 09		. 65	:	:	و :	54	:	:	:		:	۰ •	92	103	8 <.2	2 3.58	52	85 1.91	6.96 2.	31	1.18 5.	7.92 1.	17.1 7	52.0	OCA-15
NA .		620	. 095	Λ.	. 85	-	-	12 "	20	•	÷	=			7	78	53	4.54	9 4.54	4 79	24 1.24	8.20 3.	3.49 8.	2.10 3	6.23 2	19.1 6	50.0	OCA-13
NA .	725	370		. 15	. 7	:	•	11 "	60	=	-	• •	-		•	90	109	0.22	8 4.10	6 58	3.45 1.26	7.68 3.	5.68 7.	60	7.79 1.	17.0 7	50.7	OCA-12
Plag., Mte., Augite, Chlorite?		450	•	. 15	. 7	:	-	10	50	=	2.	=	. :	:	7	90	99	8.34	7 2.58	5. 37	3.15 1.65	7.98 3.	4.79 7.	11	7.38 1.	17.4 7	52.3	OCA-11
Plag., Qtz., Chl., Magnetite, Hematite		490	. 09	. 36	. 65	:		•	13	:	2 .	:		•	7	95	46	6 1.33	2 3.06	2 102	4.00 2.12	5.57 4.	2.35 5.	60	5.56 1.	17.7 5	57.0	OCA-10
Plag., Qtz., Amphibole, Chl., Mte., Augite, Calcite		420	:	. 3	.6	-	:	7 .	30	:	. =	:	=	ţ	7	138	35	5 1.70	8 2.95	2 198	3. 58 2. 52	5.78 3.	4.17 5	3.53 4	5.72 3	18.2	52.7	OCA- 9
NA .			. 095	. 27	. 6		•		38	:			-		S.	102	8 77	5 1 .78	3 3. 51	9 483	.86 2.49	. 54 3.	4.89 6.	3.84 4	4.92 3	18.5	50.7	OCA- 8
Kaolinite?, Jyroxene? NA		470			. 65	:		œ :	150	-		-	-	:	10	175	9 32	19.39	2 3.79	9 182	2.31 2.09	7.82 2	8.08 7	5.07 8	4.78 5	14.7	49.4	OCA- 7
Plag., Amphibole, Mie., Chl.,			:	u	.7	:		و. -	100	:	 •		:	:	•	89	2 126)0 . 3Z	7 2.90	1 427	2.43 1.21	8.62 2	6.81 8	4.98 6	4.06 4	15.7	50.9	OCA- 6
		245		:	.7	:		و :	48 ·	=	-	:		:	11	85	8 111	08 1.38	4 4.08	.792 54	4.23.7	6.88 4	5.08 6	68	5.95 2.	18.1	49.3	OCA- 5
NA			. 085		4.65	< 20 < 4	< 10 <	= ~	26	2 v 0	00 < 200	1 < 200	200 < 1	<1 < 2	6		7 70	36 1.67	63 3.36		4. 12 1. 12	64	4.52 7.	2.46	7.13 2	19.1 .	48.9	OCA- 4
Plass, Otz., Amphibole, Mte., Chl.,		230		^ .1	+ .7	< 20 < 4	10	7 ~	34	0 < 2	0 < 200	1 < 200	200 < 1	~	7 <	16	1 59	37 .61	4 2.37	747 174	3.45 .7	8.17 3	5.34 8	4.24	5.26 4	13.9	51.4	0CA- 3
NA		270	•	A.	4.65	< 20 < 4	< 10 <	10, ^	20	0 ~ 2	0 < 200	1 < 200	200 < 1	~	7 <	104	7 53	04 .97	98 3.04	868 9	4.61.8	8.02 4	5.12 8	4.20	5.42 4	19.4	50.4	OCA- 2
Plag., Mte., Amphibole, Chl.	550	190	:	^ .1	• • •	< 20 < 1	< 10 <	10 <	31	0 < 2	0 < 200	1 < 200	200 < 1	1 ~	5 <	118	5 63	67 .75	N	708 95	3.27 .7	73	5.96 8.	4.97	6.12 4	19.2	43.4	OCA- 1
	SrO pr	BaO p	MnO	P205	TiO2	Pt ppr	Pd pp	Mo pp	Ni ppr	Bi ppr	Te pp	Sb ppr	Ag pp	Cd pp As pp	Pb pp	Zn pp	Cu pp		н ₂ 0+	K ₂ O S ppin	Na ₂ O	CaO	МgО	FeO	Fe ₂ O	лı ⁵ 0 ³	sioz	
X-RAY ID OF MINERAL IN >2.8 gm/cc FRACTION	m	pm	•		-				n (n	~			•		m	m								3			
	- '	•.			:	VSIS	SPECTROOR A PHIC ANAL VSIS	PHIC	A 8000	PECTR	N .						÷			LYSIS	WET CHEMICAL ANALYSIS	HEMIC	WET CI					

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- AB. REPORT NO. _



130 Pemberton Ave., North Vancouver, B.C., Canada V7P 2R5 Phone (604) 985-0681 Telex 04-352667

SAMPLE SHIPMENT NOTICE

Please analyze	by	Assay (% ore g		prepared samples	
		Geochemical (ppm, trace	level)	unprepared	~
Comments C. F. M. 88.	.576 49	301-49342 -	20HN - Fle	ase analyse these	<u> </u>
	1	1 too Au	-33 Using	Teutron activat	TOR.
	- 1	1.1	ach somple	S ASTRE LLEBING	
Samples	s requir	e S.E.M. evalu	ation of	F.M. When Capl.	10 q y
after C	ooling. Se	end unused sa	Numbers	Elements to be Analyzed	araft Asiantina 217-22-0048 274
Type of Samples	No. of Samples	1	ries)	Elements to be Analyzed	
	42	49301-493.	42	Flu+ 33	
			<u></u>		
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			.1		
-				4	
				- 100	Prepaid 🞜 '
The New Complete		Date Shipp	bed June 1	3/88	Collect
Total No. Samples					
No. Parcels in Shipment					
No. Parcels in Shipment	PLE	ASE CHE	CK 🛩 BE	ELOW	
No. Parcels in Shipment	PLE	ASE CHE	•	ELOW	
REJECTS	PLE	ASE CHE	CK 🛩 BE		sly 🗆
REJECTS (COARSE OVERSIZE)	Retu	rn immediately 🛛	PULPS	Return Immediate	
REJECTS (COARSE OVERSIZE)	Retu e 90 Days - Then	rn immediately 🛛 Return C.O.D. 🗍	PULPS		
REJECTS (COARSE OVERSIZE) Store	Retu e 90 Days - Then Store 90 Days	rn immediately	PULPS	Return Immediate	D. 🗆
REJECTS (COARSE OVERSIZE) Store After 90 Days - Rejects S	Retu e 90 Days - Then Store 90 Days tored to Year End	rn immediately Return C.O.D. - Then Discard d (.35¢ Sample)	PULPS	Return Immediate Store to Year End - Then Return C.O. Store to Year End - Then Disca	D. 🗆
REJECTS (COARSE OVERSIZE) Store After 90 Days - Rejects S Results and I	Retu e 90 Days - Then Store 90 Days stored to Year End Invoices To Be S	rn immediately Return C.O.D. - Then Discard d (.35¢ Sample) Sent To	PULPS	Return Immediate Store to Year End - Then Return C.O. Store to Year End - Then Disca C. F. MINERALS RESEARC	D. 🗆
REJECTS (COARSE OVERSIZE) Store After 90 Days - Rejects S Results and I	Retu e 90 Days - Then Store 90 Days stored to Year End Invoices To Be S	rn immediately Return C.O.D. - Then Discard d (.35¢ Sample) Sent To	PULPS	Return Immediate Store to Year End - Then Return C.O. Store to Year End - Then Disca C. F. MINERALS RESEARC Id by	D.
REJECTS (COARSE OVERSIZE) Store After 90 Days - Rejects S Results and I Westmont	Retu e 90 Days - Then Store 90 Days stored to Year End Invoices To Be S Mining	rn immediately Return C.O.D. - Then Discard d (.35¢ Sample) Sent To <u>Trc</u> . A Results	PULPS	Return Immediate Store to Year End - Then Return C.O. Store to Year End - Then Disca C. F. MINERALS RESEARC	D.
REJECTS (COARSE OVERSIZE) Store After 90 Days - Rejects S <u>Results and I</u> <u>Westmont</u> 2341 South	Retu e 90 Days - Then Store 90 Days stored to Year End Invoices To Be S Mining	rn immediately Return C.O.D. - Then Discard d (.35¢ Sample) Sent To <u>Trc</u> . A Results	PULPS	Return Immediate Store to Year End - Then Return C.O. Store to Year End - Then Disca C. F. MINERALS RESEARC Id by	D.
REJECTS (COARSE OVERSIZE) Store After 90 Days - Rejects S Results and I <u>Westmont</u> 2341 South Suite 12	Retu e 90 Days - Then Store 90 Days stored to Year End Invoices To Be S Mining	rn Immediately Return C.O.D. - Then Discard d (.35¢ Sample) Sent To <u>Trcc</u> . A Results Invoices	PULPS	Return Immediate Store to Year End - Then Return C.O. Store to Year End - Then Disca C. F. MINERALS RESEARC Id by	D.
REJECTS (COARSE OVERSIZE) Store After 90 Days - Rejects S <u>Results and I</u> <u>Westmont</u> 2341 South	Retu e 90 Days - Then Store 90 Days stored to Year End Invoices To Be S Mining	rn immediately Return C.O.D. - Then Discard d (.35¢ Sample) Sent To <u>Invoices</u> Results Results	PULPS Samples submitte Shipment number	Return Immediate Store to Year End - Then Return C.O. Store to Year End - Then Disca C. F. MINERALS RESEARC d by	D.
REJECTS (COARSE OVERSIZE) Store After 90 Days - Rejects S Results and I Westmont 2341 South Suite 12 Tucson, AZ 85713	Retur e 90 Days - Then Store 90 Days Stored to Year End Invoices To Be S Mining Fricbus	rn Immediately Return C.O.D. - Then Discard d (.35¢ Sample) Sent To <u>Trcc</u> . A Results Invoices	PULPS Samples submitte Shipment number	Return Immediate Store to Year End - Then Return C.O. Store to Year End - Then Disca C. F. MINERALS RESEARC Id by	D.
REJECTS (COARSE OVERSIZE) Store After 90 Days - Rejects S Results and I Westmont 2341 South Suite 12 Tucson, AZ 85713 HTTM: H.Dum	Return e 90 Days - Then Store 90 Days Stored to Year End Invoices To Be S Mining Fricbus	rn immediately Return C.O.D. - Then Discard d (.35¢ Sample) Cent To Circe. A Results Invoices Results Invoices	PULPS Samples submitte Shipment number	Return Immediate Store to Year End - Then Return C.O. Store to Year End - Then Disca C. F. MINERALS RESEARC d by	D.
REJECTS (COARSE OVERSIZE) Store After 90 Days - Rejects S Results and I Westmont 2341 South Suite 12 Tucson, AZ 85713 7+TTM: H.Dum C. F. MINERALS RESU	Return e 90 Days - Then Store 90 Days stored to Year End Invoices To Be S Mining Fricbus	rn immediately Return C.O.D. - Then Discard d (.35¢ Sample) Cent To Circe. A Results Invoices Results Invoices	PULPS Samples submitte Shipment number Client project num	Return Immediate Store to Year End - Then Return C.O. Store to Year End - Then Disca C. F. MINERALS RESEARC Ind by	D.
REJECTS (COARSE OVERSIZE) Store After 90 Days - Rejects S Results and I Westmont 2341 South Suite 12 Tucson, AZ 85713	Return e 90 Days - Then Store 90 Days itored to Year End Invoices To Be S Mining Fricbus mett EARCH LTD.	rn immediately Return C.O.D. - Then Discard d (.35¢ Sample) Cent To Circe. A Results Invoices Results Invoices	PULPS Samples submitte Shipment number Client project num	Return Immediate Store to Year End - Then Return C.O. Store to Year End - Then Disca C. F. MINERALS RESEARC d by	D.

•	•	and a second sec	•	AB. REPORT NO.
B		R-CLEG	<u>3 & COI</u>	MPANY LTD.
130 Pemberton Ave.	, North Vanc	ouver, B.C., Canada	V7P 2R5 Phone (604) 985-0681 Telex 04-352667
		SAMPLE SHIPM	ENT NOTICE	
Please analyze Comments <i>C. F.M. 88</i>	e by - 576 49 	Assay (% ore gr Geochemical (ppm, trace) 7301 - 49342 halyse these		prepared samples unprepared <i>gm</i>) - Please geochem <i>ed Samples for</i> ods.
	<u></u>		/	
Type of Samples	No. of Samples	Sample N (Ser		Elements to be Analyzed
	42	49301 - 4934	22	Cu-Hg
			· · · · · · · · · · · · · · · · · · ·	
· · · ·			.1	
				4
		Date Shipp	ed June 13/	188 Prepaid 12 Collect
Total No. Samples No. Parcels in Shipment	PLE	ASE CHE		
REJECTS			PULPS	Return Immediately
(COARSE OVERSIZE	noru	n immediately		Return miniousier,
Sto	re 90 Days - Then Store 90 Days	Return C.O.D. 🛛	Sto	ore to Year End - Then Retain C.C.D.
After 90 Days - Rejects				Store to Year End - Then Disourd
Results and	Invoices To Be S	ent To		C. F. MINERALS RESEARCH LTD.
Westmont 1	Mining I	nc. Results	Samples submitted b	KELOWNA, B.C. VIY 5W6
<u>2341 South 1</u> Suite 12	Friebus	X Invoices	Shipment number	
TURSON AZ		Results		
85713 Atta: H. Dun	nmett		Client project numbe	r
C.F. MINERALS R 263 LAKE KELOWNA, B.C	ESEARCH LT	D. Ø Results	Purchase order num	nber

C.F.MINERAL RESEARCH LTD. 263 LAKE AVENUE KELOWNA, BRITISH COLUMBIA CANADA V1Y 5W6

C.F.M. 88-576

WESTMONT PROJECT: H. DUMMETT 13/06/88

SAMPLE NUMBER	ORIGINAL WEIGHT (KG)	FRACTION	WEIGHT (GMS)
49301	9.000		
49301	51000	-20HM	73.91
49301		-20HP	185.98
49301		-20HN	81.65
49302	9.200		
49302		-20HM	91.03
49302		-20HP	156.23
49302		-20HN	152.15
49303	10.800		
49303		-20HM	81.57
49303		-20HP	196.30
49303		-20HN	102.58
49304	8.300		
49304		-20HM	22.52
49304		-20HP	149.28
49304		-20HN	6.64
49305	6.700		
49305		-20HM	26.84
49305		-20HP	134.56
49305		-20HN	53.46
49306	8.000		
49306		-20HM	17.29
49306		-20HP	74.36
49306		-20HN	3.47
49307	7.900		
49307		-20HM	9.27
49307		-20HP	46.62
49307		-20HN	3.27
49308	7.500		
49308		-20HM	14.22
49308		-20HP	53.88
49308		-20HN	5.10

C.F.M. 88-576

WESTMONT

SAMPLE NUMBER	ORIGINAL WEIGHT (KG)	FRACTION	WEIGHT (GMS)
49309	6.500		
49309	0.300	-20HM	9.49
49309		-20HP	21.49
49309		-20HN	3.77
49310	11.600		
49310		-20HM	10.95
49310		-20HP	36.88
49310		-20HN	5.59
49311	8.100		
49311		-20HM	1.81
49311		-20HP	32.40
49311		-20HN	4.16
49312	7.600		
49312		-20HM	3.20
49312		-20HP	19.92
49312		-20HN	2.26
49313	7.400		
49313		-20HM	38.79
49313		-20HP	60.92
49313		-20HN	13.59
49314	7.100		
49314		-20HM	427.54
49314		-20HP	508.40
49314		-20HN	84.89
49315	9.000		
49315		-20HM	763.66
49315		-20HP	605.50
49315		-20HN	172.74
49316	10.600		
49316		-20HM	2182.55
49316		-20HP	651.17
49316		-20HN	32.35

C.F.M. 88-576

WESTMONT

SAMPLE NUMBER	ORIGINAL WEIGHT (KG)	FRACTION	WEIGHT (GMS)
49317	10.800		410 ATO 500 ATO 400 ATO
49317	10.000	-20HM	782.67
49317		-20HP	864.80
49317		-20HN	98.76
49318	7.500		
49318		-20HM	8.93
49318		-20HP	43.69
49318		-20HN	17.29
49319	7.100		
49319		-20HM	9.87
49319		-20HP	32.89
49319		-20HN	2.59
49320	9.400		
49320		-20HM	521.52
49320		-20HP	418.67
49320		-20HN	62.91
49321	9.300		
49321		-20HM	124.90
49321		-20HP	215.03
49321		-20HN	22.61
49322	9.500		
49322		-20HM	746.06
49322		-20HP	488.27
49322		-20HN	87.86
49323	6.600		
49323		-20HM	7.81
49323		-20HP	17.04
49323		-20HN	2.07
49324	7.600		
49324		-20HM	23.75
49324		-20HP	27.56
49324		-20HN	18.35

WESTMONT

SAMPLE NUMBER	ORIGINAL WEIGHT (KG)	FRACTION	WEIGHT (GMS)
49325	8.200		
49325	0.200	-20HM	155.25
49325		-20HP	488.24
49325		-20HN	11.11
49326	8.100		
49326		-20HM	8.46
49326		-20HP	9.93
49326		-20HN	1.38
49327	8.000		
49327		-20HM	9.63
49327		-20HP	10.61
49327		-20HN	2.03
49328	9.000		
49328		-20HM	1.35
49328		-20HP	18.63
49328		-20HN	2.05
49329	8.200		
49329		-20HM	3.00
49329		-20HP	13.00
49329		-20HN	4.30
49330	8.100		
49330		–20HM	6.69
49330		-20HP	21.72
49330		-20HN	3.02
49331	8.200		
49331		-20HM	2.24
49331		-20HP	14.21
49331		-20HN	1.72
49332	6.500		
49332		-20HM	0.16
49332		-20HP	188.20
49332		-20HN	56.96

WESTMONT

SAMPLE NUMBER	ORIGINAL WEIGHT (KG)	FRACTION	WEIGHT (GMS)
49333	6.700		
49333	01100	-20HM	27.67
49333		-20HP	95.70
49333		-20HN	16.01
49334	7.700		
49334		-20HM	51.33
49334		-20HP	254.12
49334		-20HN	33.45
49335	6.200		
49335		-20HM	2.13
49335		-20HP	17.53
49335		-20HN	3.77
49336	8.100		
49336		-20HM	49.11
49336		-20HP	148.44
49336		-20HN	95.69
49337	6.200		
49337		-20HM	58.11
49337		-20HP	227.01
49337		-20HN	24.35
49338	5.600		
49338		-20HM	23.80
49338		-20HP	157.29
49338		-20HN	18.09
49339	6.600		
49339		-20HM	139.58
49339		-20HP	271.46
49339		-20HN	66.23
49340	9.500		
49340		-20HM	148.13
49340		-20HP	205.88
49340		-20HN	30.60

WESTMONT

H. DUMMETT 13/06/88

SAMPLE NUMBER	ORIGINAL WEIGHT (KG)	FRACTION	WEIGHT (GMS)
49341	10.000		
49341	10.000	-20HM	104.09
49341		-20HP	401.45
49341		-20HN	66.92
49342	7,600		
49342		-20HM	69.82
49342		-20HP	186.78
49342		-20HN	21.39

ALL SAMPLES HAVE BEEN UV LIGHT EXAMINED-NO SCHEELITE GRAINS WERE FOUND.

Little Suld Sulch Area

Figure 2 - LGG Geochemical Sampling Results

Normalized 103 tatios

-	Vein/dump	- (ppm)				6	1 M5	r 81	IZn	1 Ag	Au
•	<u>Cu</u>	<u>Pb</u>	Zn	Ag	Λu	Cu Mo	1013			1.0	
:1001	1400	40	160	6.8	.71	· · ·	10.23	7.85	8.41	917	8.05
2	1450	130	165	7.8	1.0		1 1 1 1 1 1 1 1 1 1 1 1 1 1 1 1 1 1 1	1	i	907	
3	1000	190	245	18.0	3.50	2 9.65	i	4	8.57	9.31	8.34
5	210	90	120	5.4	8.20	1.00			8.41	9.36	8.57
6	175	950	4300	12.0	.15	- 8.91		8.52			862
7	1600	2600	3700	350	.67	4 8.22 2 8.57	9.78	8.99		9.10	7.15
ક	445	5400	6600	50.0	1.20	2 8.57 2 8.37	8.67.			10.54	7./7
(-1	350	6600	18500	72.0	6.0	0	1		9.71	9.54	7.79
15	395	50	1150	24.0	.13	2 8.00			10.04		8.24
175	120	3600	3100	350.	1.3	4 7.47		6.08	7.44	7.77	5.49
25	205	4500	2900	51.0	.95	4 7,42 6 8.04	1	8.93 9.49		10.50	7.45
	345	10500	2700	140.0	.56	24 7.83	:	9.41		9.57 9.57	7.71 7.04
.36	380	4400	12500	39.0	.18	8 8.12	9.58			9,19	6.79
41	520	12500	3500	350.	e . 19	4 7.98	8.89				6.53
42	190	105	305	8.8	.04	4 8.55	10.43		0.07 8.77		6.86
43	235	1500	950	140.0	.53	22 7.79	10.12			9.75	
41	295	12500	470	350.	.15	8 7.72		9.46			6.43
+5	255	1860	1650	54.0	. 54	8 8.18		9.08			7.49
46	1060	5600	1150	130.0	7.65	55 8.15	10.41				8.00
47	(13800)	360	480	9.3	.22	3 10.46				8.72	7.07 20
÷ 7	240	5400	1300	83.0	3.19	8 8.01		9.47	8.77		8.14
50	610	11600	2900	62.0	1.95	4 8.39	9,27	9.92	211	9.51	7.89
57	310	7000	5500	83.0	.55	8 8.03			9.37	9.60	7.28
. 54	116	500	1300	120.0	1.45	3 7.84				10.37	7.93
57	52	950	7000	41.0	.64	13 7.32	10.03				7.4/
58	320	4600	9000	170.0	7.35	3 7.97	8.98			9.98	8.34
51	76	2040	880	27.0	1.63	5 7.84	9.93				
					<u> </u>	-	8.58	R.91	892	9.14	
					$(n=2i)\overline{x}$	8.25 0.72	0.80	0.74	0.69	0.70	
					x+20	9.69	10.18	10.39	k). 31	10.54	
	X=969/475	3910	5360	102.0	1.87			10.01	10.01	10.01	
	s=2601/443	4016	8041	115.12	2.48					•	
	Silicified-	-brecciated	zones								
	110	-	15	-	-	12					
	45	_ ·	25	-	-	4					
	60		20	-		18					
	100	185	40	.2	-	20					
	115	30	15	2.6	.07	44 (mass	.jar.)			
	65	40	55	.2	-	14					
	120	-	65	.2	.02	6					
											
	X=88	37	34	.46	.02						
	s=30	67	20	.95	.02						

SADDLE MTTN. PLUTONIC SUITE

SAODLE MITN.

33 a

DATA SHEET FOR CHEMICAL AND AGE DATA

15

FOR IGNEOUS COMPLEXES

LOCATION: SADDLE MOUNTAIN, PINAL COUNTY ARIZONA

LOCATION COORDINATES:

K₅₅ : 1.00 K_{57.5}: 1.3 1.30 CA

к₆₀ :1.6]155

Fet3/Fet2: 1.69 (oxidized) KCa: 68.5 CA KMg: 63.0 CA Fe: poor A1: 0.930-1.015 (mostly metalum.)

PEACOCK INDEX: Affinity: retaluminous to useally recaluminous, calc-alkalic, iron-poor.

ROCKS INVOLVED: Microdiorite dikes (~70 m.y.?), Ash Creek stock (62m.) Narious dikes (62-70 m.y.)

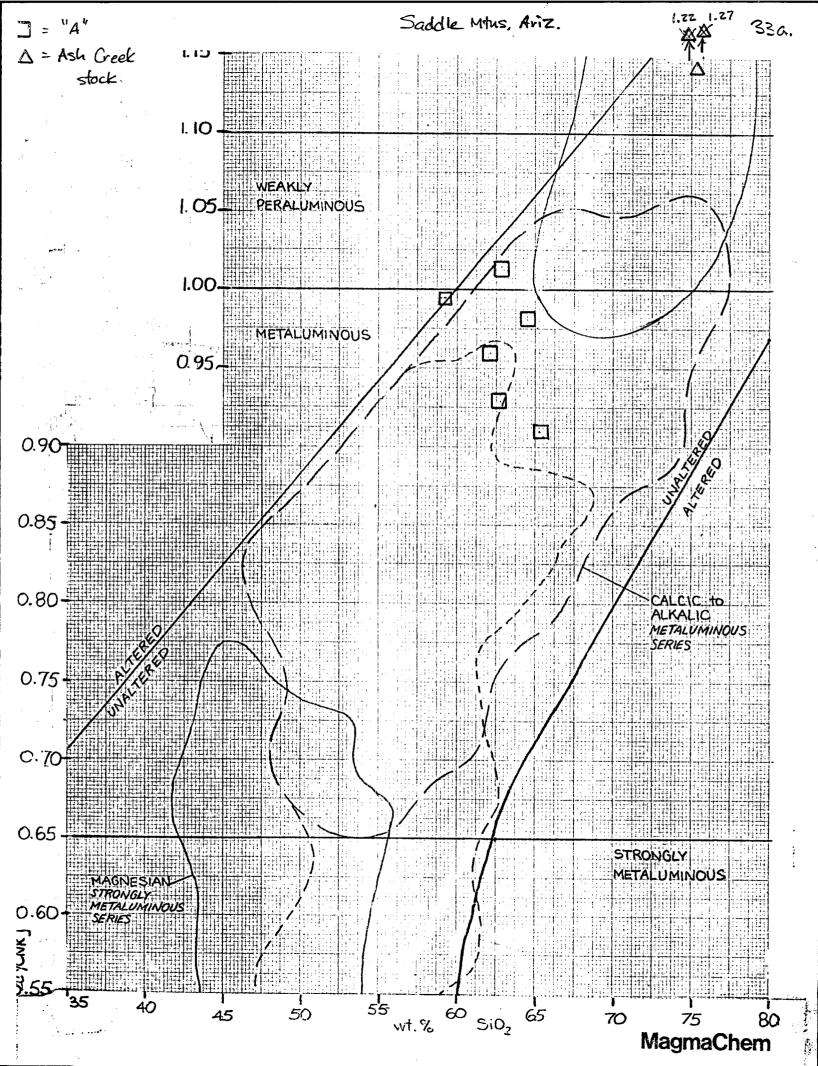
AGE DATA: 70-61 M.V.

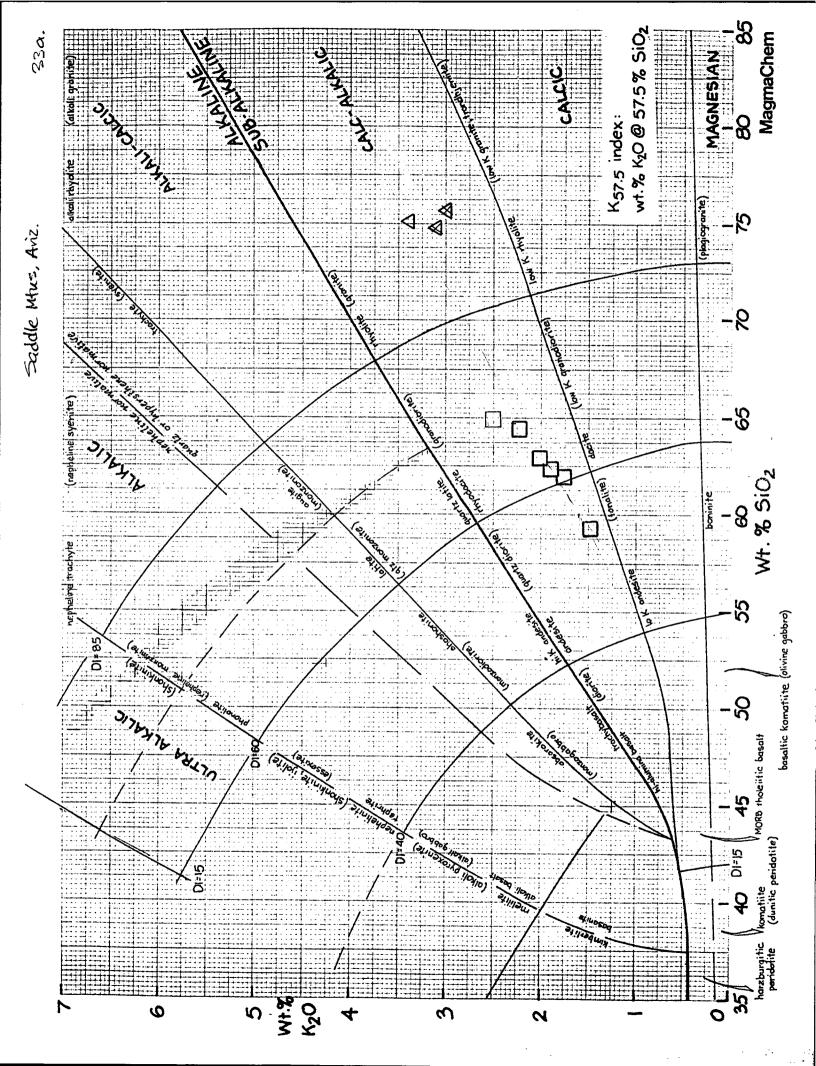
COMMENTS;

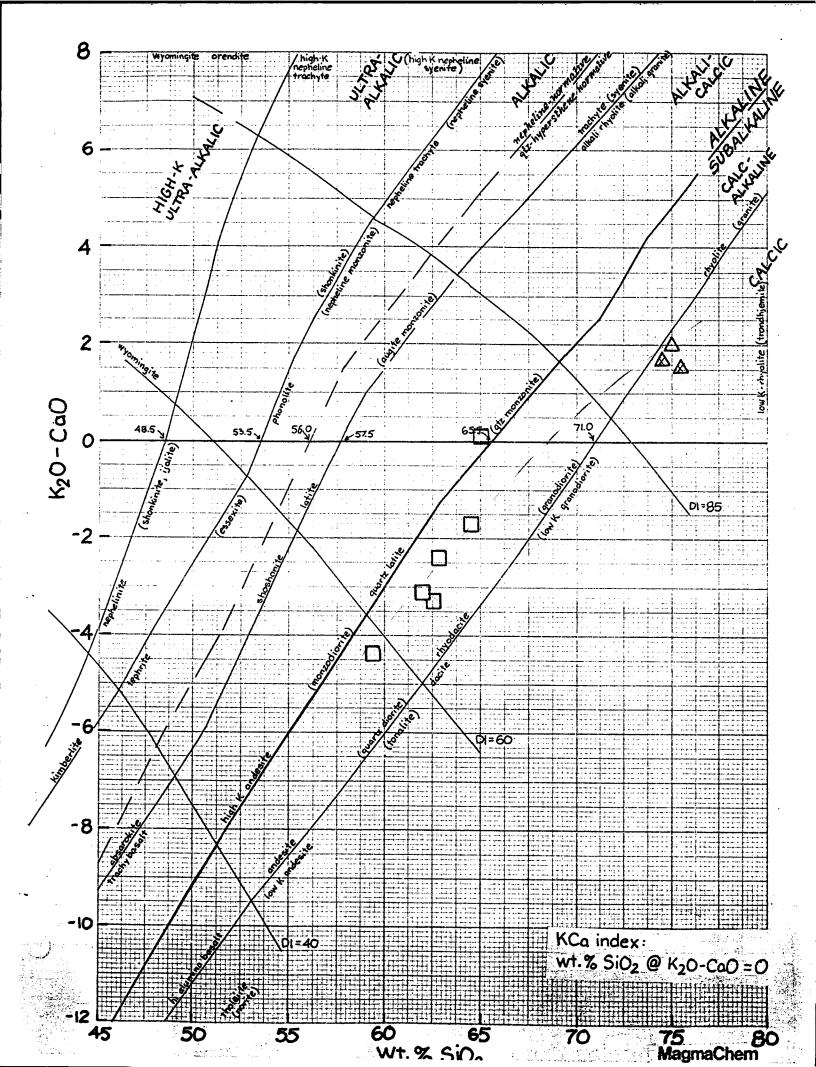
MORENCI Assemblage

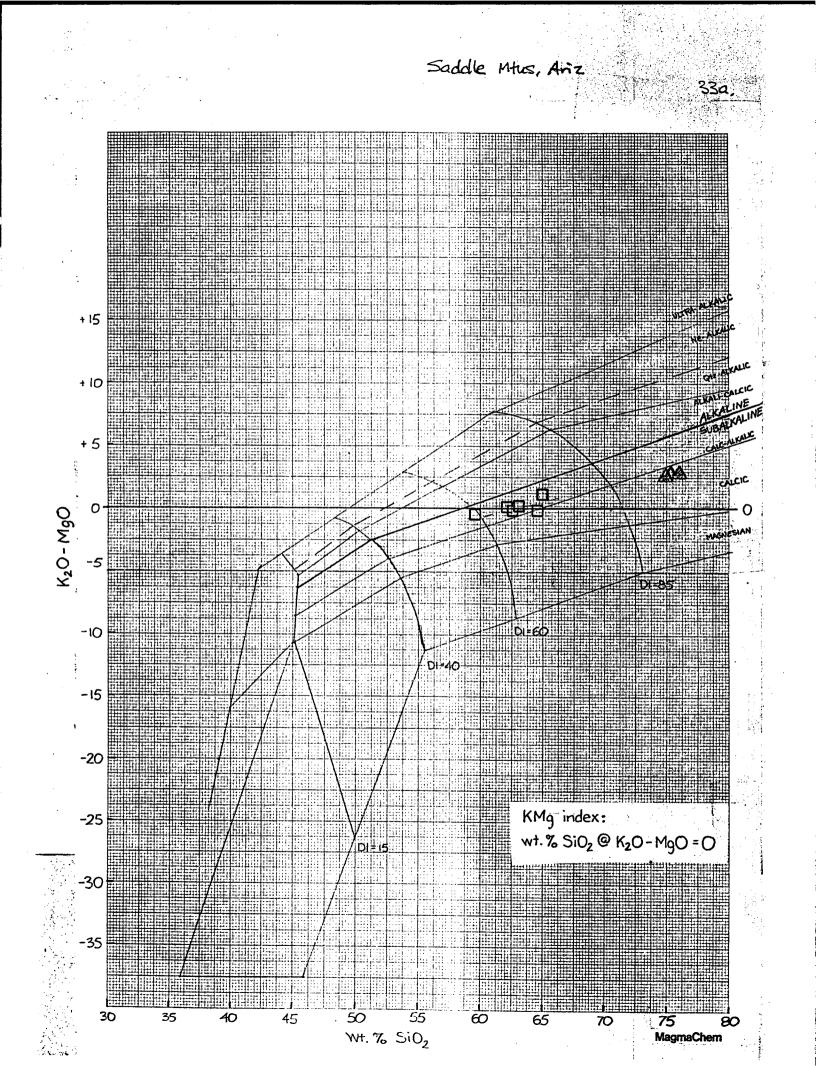
REFERENCES FOR CHEMICAL DATA: Keith, Renrig, Barrett, Clayton Unpub. data).

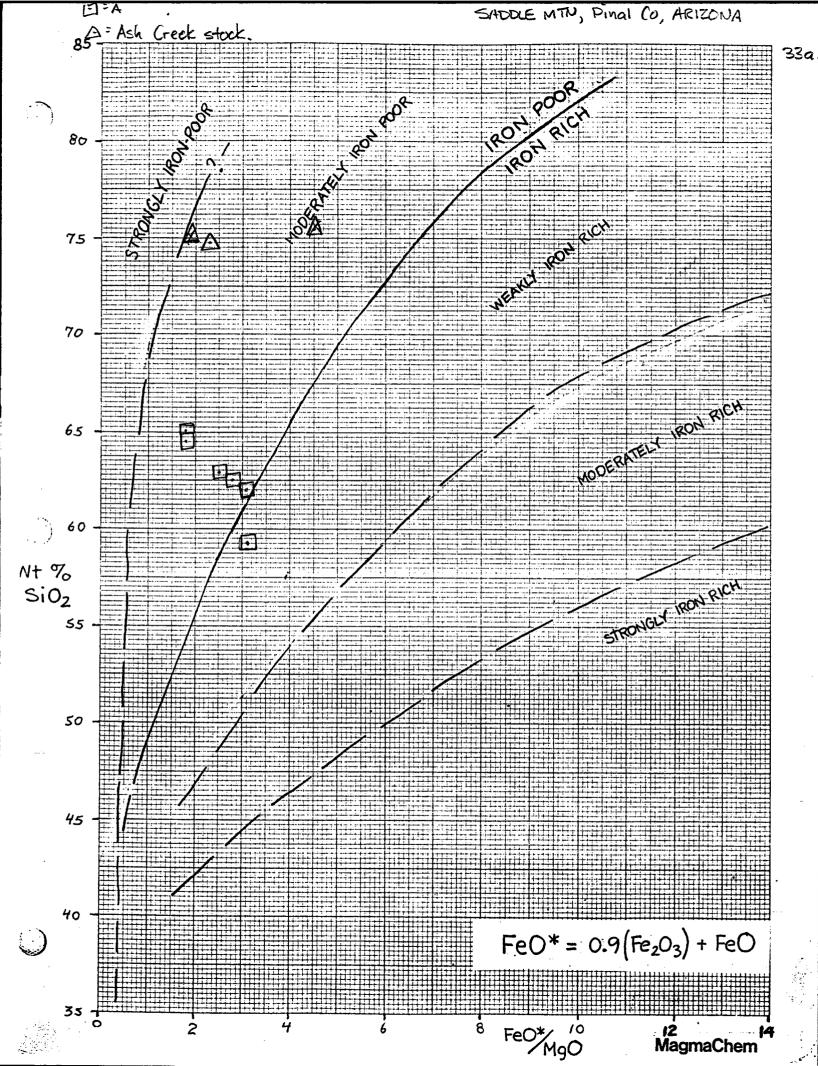
REFERENCES FOR AGE DATA: Keith and Damon (univer data).

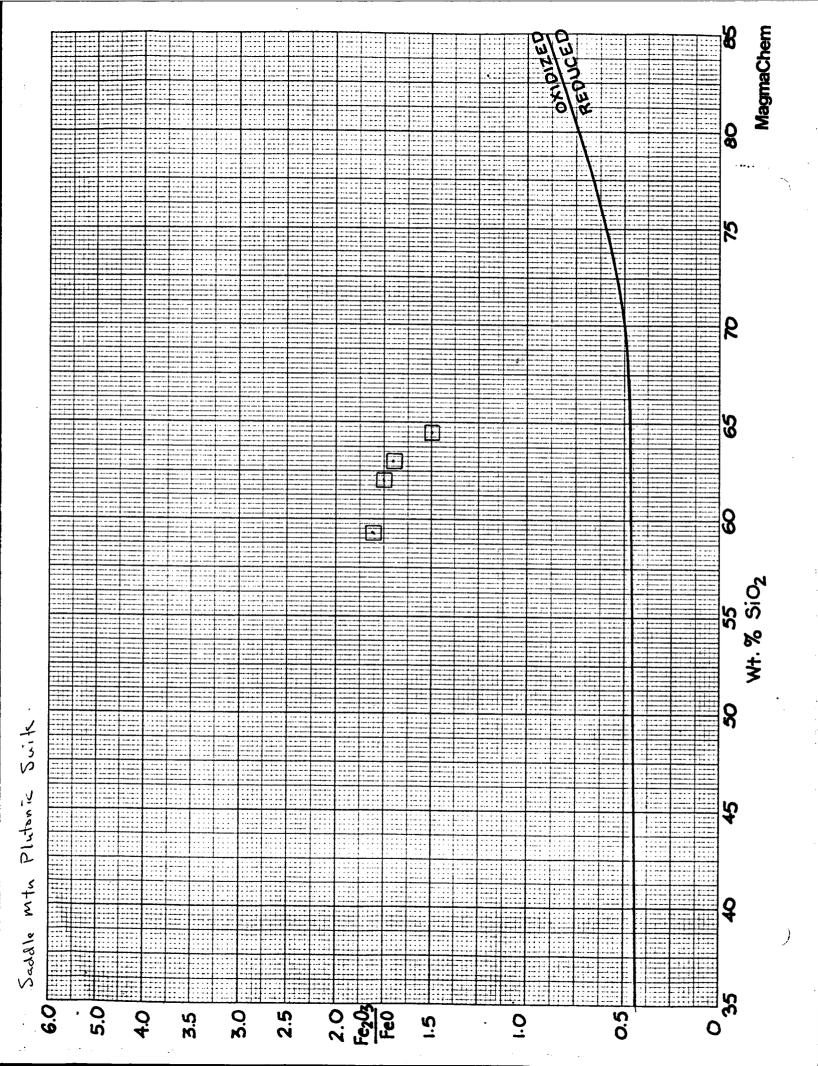


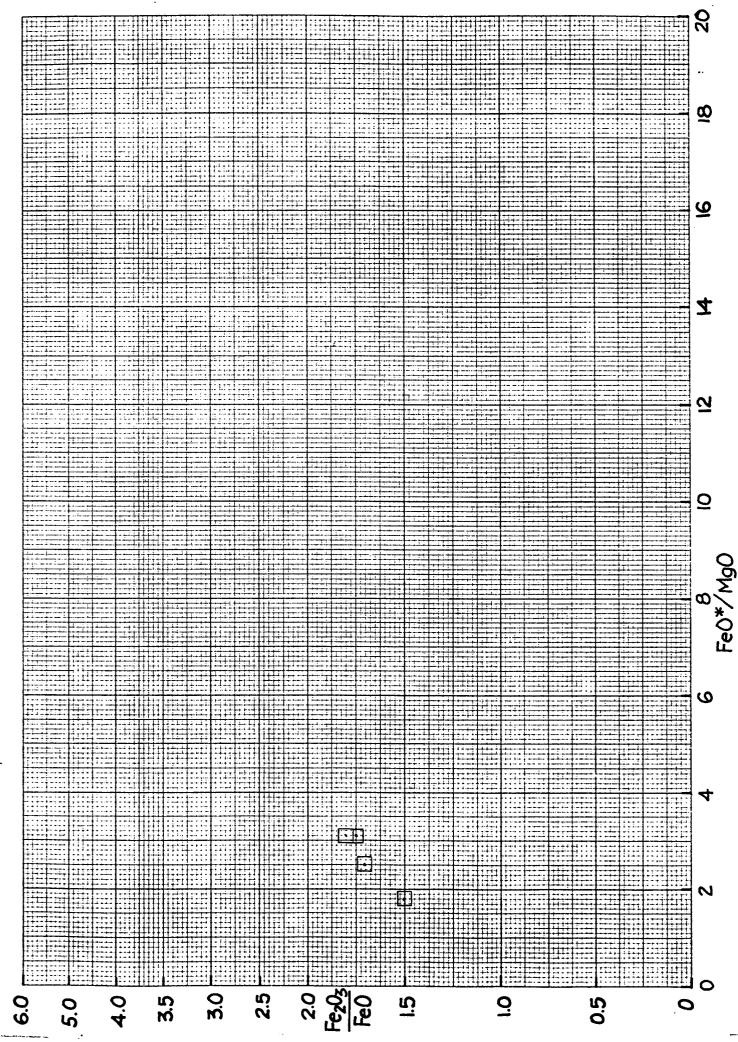




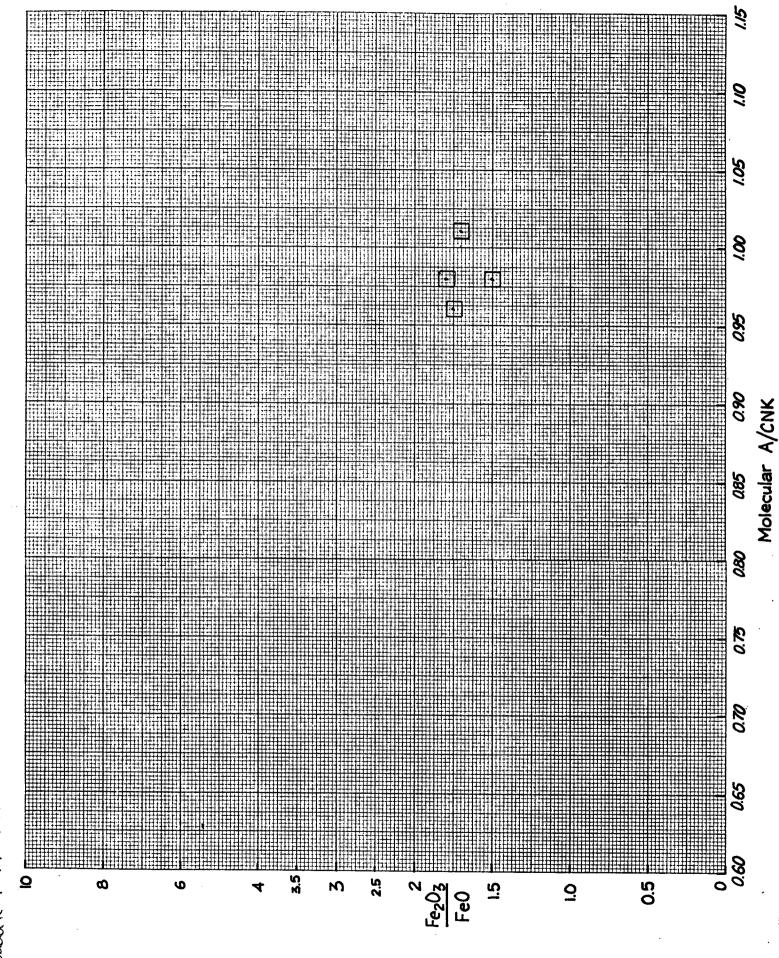




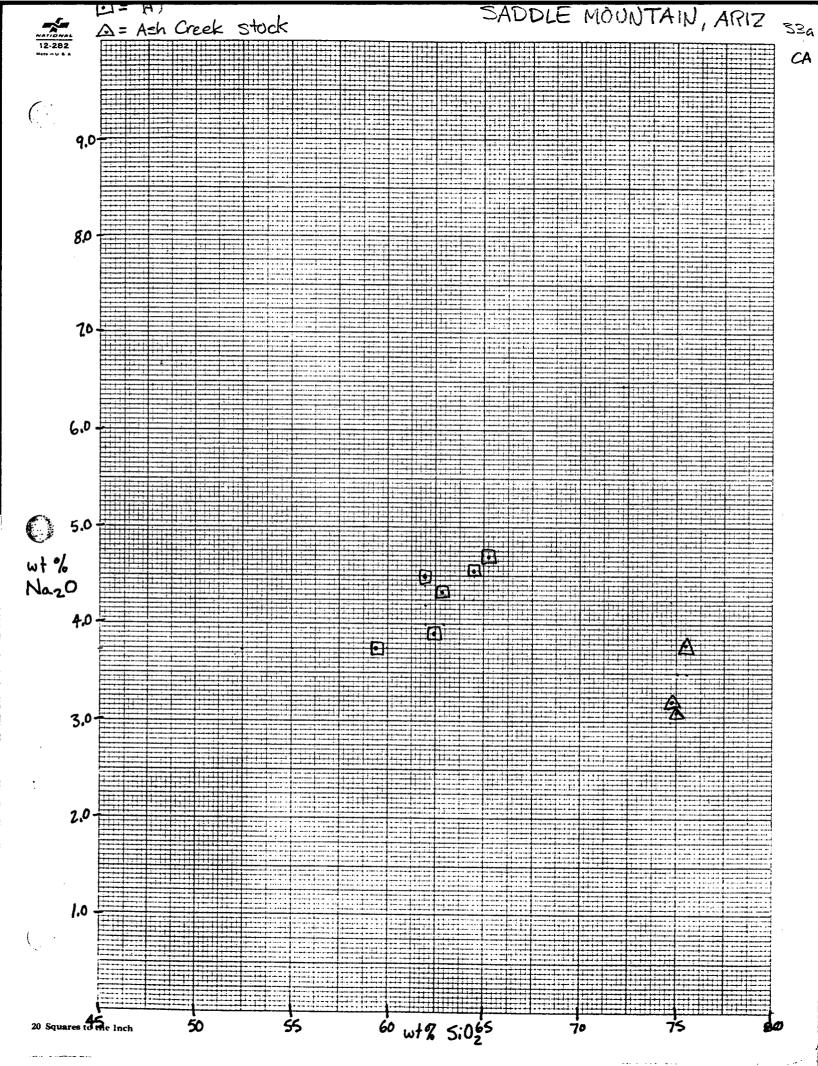


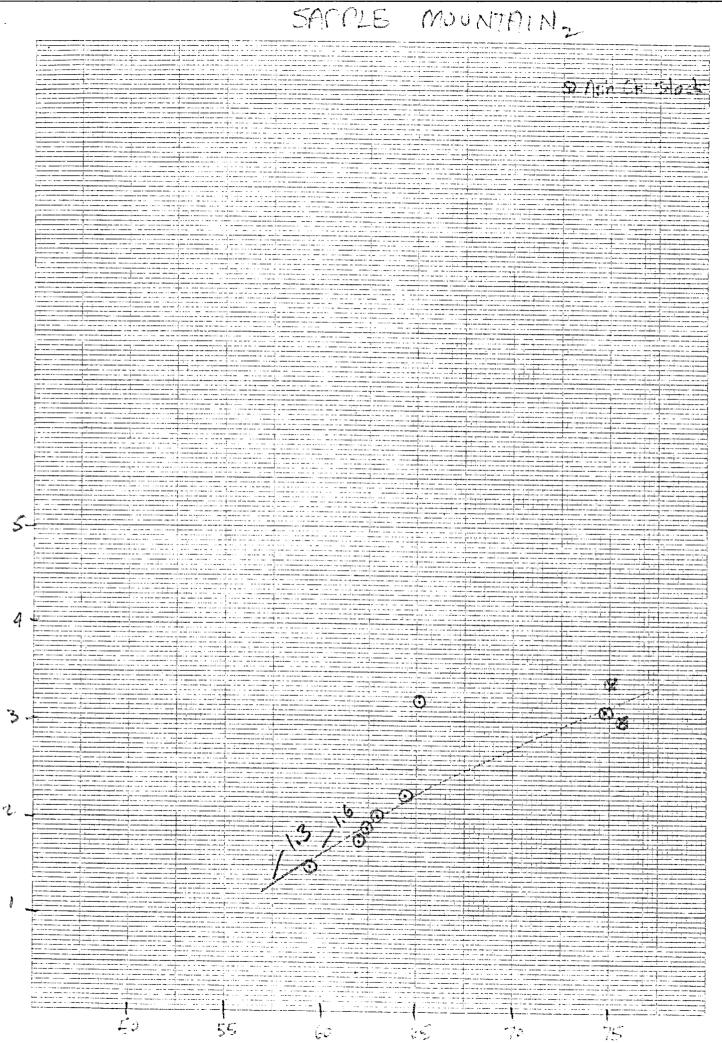


suite pictoric いたい Sac'd le

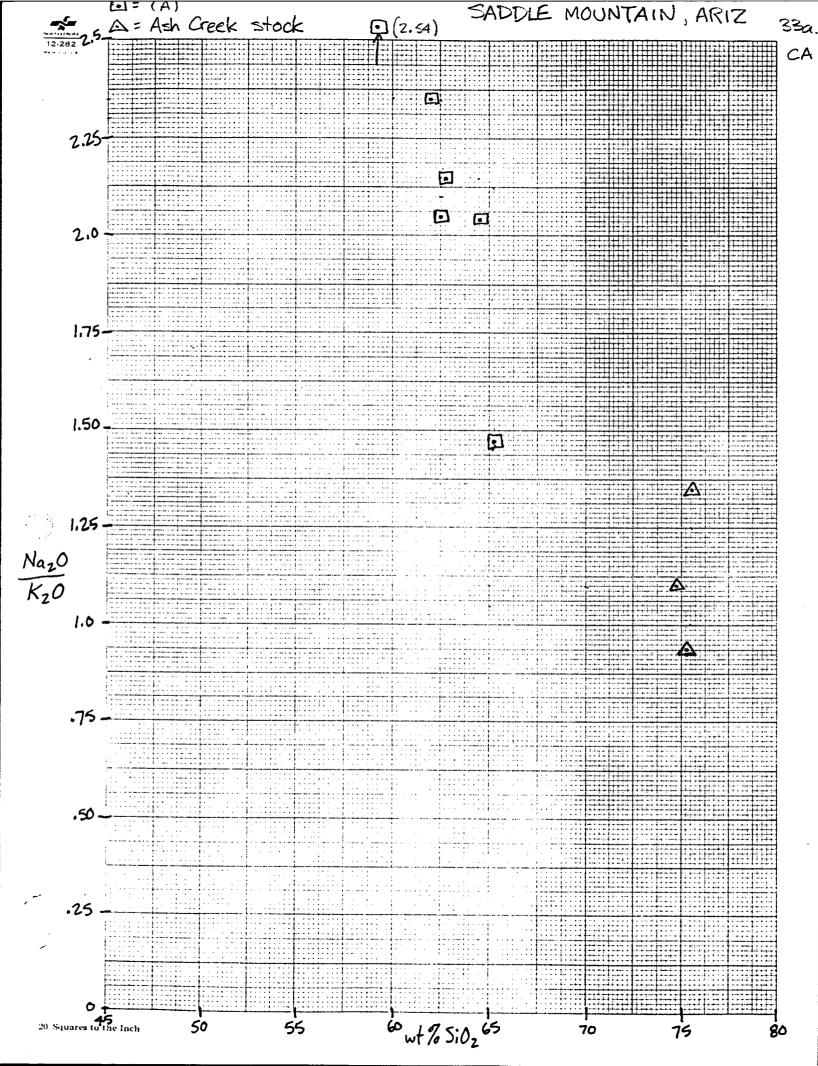


Saddle Mtn Plutanic Suite





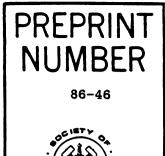
CEFFICIENCYO 22-107



				SAD	PLE I	MOUN	TAIN		332
	A		r.		Å.	1 5	-	B	P (7 C)
Sample	T-460	·	19 sam	-2 T-32	054 T-4	59 4 5AI	n-3 /f	1 C-3 A	C-2 AC-3
5:02	59,3	62.					1		,08 75.60
Al203	184	17,					1		4 13,0
Fe2 03	4.22	3,3						a4 .91	•
FeO	2,33	1.89		1,59			•		
MnO			,07			,05	,0:	3,02	,02
1	1,95	1,57		1.62	2,33		i		
Mgo GO		4,87			-	2,0	,37		
	5,84					3,)	1.4	1.4	1,4
Nazo	3.74	4.14	3,9	4,32		4.7	3.4	3,2	3,8
K20 -	1.47	1.76	1.9	2,01	2,23	3.2	3.1	3,4	3,0
P2Os		~					J vr	45	
	0.6	0,5	,59	-1	0,50	.52	.14	.13	.12
62	.39	.56	20	.56	2.2		.		
H_{20} +	2,27	1,55	,39	2.05	1.83	.51	1.0	1.)	.45
H20-			1, 88						$\Delta_{\rm e}$
	99.91	98.84	99,73	100.09	100.72	98,8	99.54	99.14	98.58
Rb						16.9	,	57.9	
57 875r/8657						490		342	
815r/865r					-	.7045		7066	
A/CNK	0.98	0-96	0.93	1.01	0.98	0.91	1.22	1.14	1.05
Na20/K20	. 2.5.4	2,35	2.05	2.15	2.04	1.47	1.10	0.94	1.27
K20 - Mg0	-0.48	0.19	-0.10	0.39	-0.10	1.20	2.73	2.95	2.80
	-4.37	-3.11	-3.30	-2.41	-1.70	0.10	1.70	2.00	(.60
FeO*/MgD	3.1	3.1	2.8	2.5	1.8	1.8	2.3	.1.9	4.5
F/F	1-81	1.75		1.71	1.49				
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SOCIETY OF MINING ENGINEERS OF AIMF

CALLER NO. D. LITTLETON. COLORADO 80127



GEOLOGY OF THE FORTITUDE GOLD-SILVER DEPOSIT,

COPPER CANYON, LANDER COUNTY, NEVADA

P. R. Wotruba

R. G. Benson

K. W. Schmidt

Battle Mountain Gold Company Battle Mountain, Nevada

For presentation of the SME Annual Meeting

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PREPRINT AVAILABILITY LIST IS PUBLISHED PERIODICALLY IN MINING ENGINEERING Abstract. The Fortitude gold-silver deposit is related to a "wallrock" copper porphyry system developed within Middle Pennsylvanian to Permian Antler Sequence rocks adjacent to a Middle-Tertiary altered granodiorite intrusive stock at Copper Canyon. Gold-silver ores of the Fortitude deposit occur with disseminated and massive sulfide replacement mineralization of skarn-like or calc-silicated limy horizons of the Antler Sequence contact meta-sedimentary rocks. A major, north-trending, steeply westward dipping normal fault was important as a conduit for hydrothermal fluids responsible for the metallization in the Fortitude area. Gold-silver mineralization is best developed near a marble front where retrograde chloritization and destruction of prograde calc-silicate mineral phases is most prevalent. Fluid-inclusion studies performed on the Copper Canyon system indicate a wide variation in fluid chemistry during several hydrosilicate stages that ranged in temperature from 500°C to 220°C.

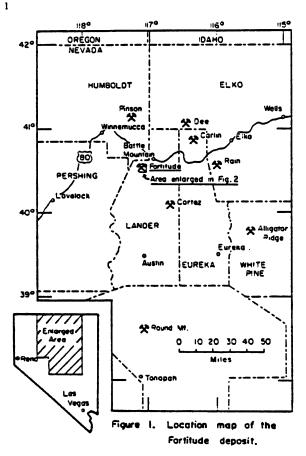
Introduction

The Fortitude deposit is located in the Battle Mountain mining district, which is in the central Great Basin region of the Basin and Range province in the western United States. The deposit is at Copper Canyon in the southernmost portion of the district, approximately 19 km southwest of the town of Battle Mountain, Lander County, Nevada (see Fig. 1).

Mining activity within the Battle Mountain district began in 1866 (Roberts and Arnold, 1965). Base and precious metals were mined by underground methods intermittently up to 1955. Placer gold was first discovered at Copper Canyon in 1912. Moderate-scale dredging of alluvial gravel at the mouth of Copper Canyon was conducted by the Natomas Company from 1944 to 1955 (Johnson, 1973). Large-scale open-pit mining of copper ore was initiated in 1967 by Duval Corporation (Sayers, Tippett, and Fields, 1968). As copper prices decreased and precious-metals prices increased in the mid-1970's, Duval Corporation's exploration efforts shifted to precious metals at Copper Canyon. These exploration efforts resulted in the discovery of the Tomboy-Minnie deposits, and shortly afterward the Northeast Extension deposit at Copper Canyon (Blake, Wotruba, and Theodore, 1984; see Fig. 2). The Fortitude deposit was discovered in late 1980, after Duval Corporation had already proceeded with precious-metals production at Copper Canvon.

Conversion of the existing plant facilities at Copper Canyon to a precious-metals recovery system was completed in 1978 (Jackson, 1982), and production from the Tomboy-Minnie deposits began in early 1979. Precious-metals production moved to the Northeast Extension deposit at Copper Canyon in 1982. In December of 1984, after three years of preproduction stripping and mining of the Upper Fortitude ore zone, production from the highergrade Lower Fortitude ore zone began. Currently, under a new corporate name of Battle Mountain Gold Company, the Fortitude deposit provides ore for continued precious-metals production at Cepper Canyon.

Recorded metal production from the Battle Mountain mining district prior to 1967 included 4666 kg (150 000 uz) of gold, 65 317 kg (2.1 million uz) of silver, 13 600 t (15 000 st) of



copper, 4500 t (5000 st) of lead, and 13 600 t (15 000 st) of zinc (Roberts and Arnold, 1965). At the time of Duval's 1967 start-up of production, ore reserves at Copper Canyon included 12.6 Mt (13.9 million st) in the East ore body and another 3.6 Mt (4.0 million st) in the West ore body, both with average grades of 0.79% copper, 16.1 g/t (0.47 oz/st) silver and .86 g/t (0.025 oz/st) gold (Blake and others, 1984). The Tomboy-Minnie deposits collectively contained 3.5 Mt (3.9 million st) of ore grading 3.09 g/t (0.09 oz/st) gold and 9.60 g/t (0.28 oz/st) silver. The Northeast Extension reserves were identified as 1.4 Mt (1.5 million st) of ore with an average grade of 2.91 g/t (0.085 oz/st) gold and 5.1 g/t (0.15 oz/st) silver. Initial total reserve estimates for the Fortitude deposit were 14.5 Mt (16 million st) grading 5.14 g/t (0.150 oz/st) gold and 29.8 g/t (0.87 oz/st) silver (Jackson, 1982). Therefore, prior to large-scale mining, the Copper Canyon area contained a minimum of 102 642 kg (3.3 million oz) of gold and 541 201 kg (17.4 million oz) of silver as minable reserves.

The Fortitude deposit reserves as of January 1, 1985 were 10 605 000 t (11.69 million st) grading 5.28 g/t (0.154 oz/st) gold and 28.49 g/t (0.83 oz/st) silver. Currently, the Battle Mountain Gold Company is producing 6800 kg (220 000 oz) of gold per year and significant amounts of silver, processing approximately 2720 to 3175 t (3000 to 3500 st) per day. Extraction of gold and silver is by the carbon-in-pulp, cyanide-leach method (Anderson and Todd, 1984).