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APPENDIX A

Calculation of weighted mean atomic number (\bar{Z}) and backscattered-electron coefficient ($\bar{\eta}$)

The backscattered-electron coefficient for pure elements can be approximated by the following empirical formula: $\eta = 0.5 - 0.228 \cdot 10^{-4} (80 - Z) \cdot (80 - Z)^{1.3}$ where Z is the atomic number of the element (Heinrich, 1981). For a multicomponent target, such as a mineral, the weighted mean backscattered-electron coefficient, $\bar{\eta}$, is used where $\bar{\eta} = \sum C_i \eta_i$ and C_i is the weight fraction of element i in the mineral, and η_i is the backscattered-electron coefficient of element i .

Weighted mean atomic number $\bar{Z} = \sum C_i Z_i$. For comparison with $\bar{\eta}$ this value can be divided by 100.

By way of illustration \bar{Z} and $\bar{\eta}$ is calculated for pyrite with composition FeS_2 (Table 1).

Table 1 Calculation of weighted mean atomic number (\bar{Z}) and backscattered-electron coefficient ($\bar{\eta}$) for pyrite with composition FeS_2 .

<i>i</i>	<i>Fe</i>	<i>S</i>
<i>Z</i>	26	16
η	0.28	0.18
C_i	0.47	0.53
$C_i \eta_i$	0.13	0.09
$\bar{\eta}$	0.22	
$C_i Z_i$	12.10	8.55
$\bar{Z}/100$	0.21	

APPENDIX B

Ideal chemical formulae and densities of minerals in UG2 chromitite (Hurlbut & Klein, 1977; Cabri, 1981; Deer et al.; 1983).

Mineral name	Ideal chemical formula	Density
Pentlandite	(Fe,Ni) ₉ S ₈	4.6-5.0
Pyrrhotite	Fe _{1-x} S	4.58-4.65
Pyrite	FeS ₂	5.02
Millerite	NiS	5.5±0.2
Chalcopyrite	CuFeS ₂	4.1-4.3
Galena	PbS	7.4-7.6
Braggite	Pt _{0.64} Pd _{0.27} Ni _{0.14} S _{1.00}	9.34
Cooperite	PtS	10.12
Vysotskite	PdS	6.71
Laurite	(Ru,Os,Ir)S ₂	6.22
Malanite	Cu(Pt,Rh) ₂ S ₄	7.4
Pt-Fe alloy	Pt ₃ Fe	18.23
Chromite	(Fe,Mg)Cr ₂ O ₄	4.6
Talc	Mg ₃ Si ₄ O ₁₀ (OH) ₂	2.7-2.8
Bronzite	(Mg,Fe)SiO ₃	3.3
Anorthite	CaAl ₂ Si ₂ O ₈	2.76
Albite	NaAlSi ₃ O ₈	2.62
Quartz	SiO ₂	2.65
Calcite	CaCO ₃	2.71
Pumpellyite	Ca ₄ (Mg,Fe)(Al,Fe) ₅ (OH) ₃ [Si ₂ O ₇] ₂ [SiO ₄] ₂ .2H ₂ O	3.18-3.23
Chlorite	(Mg,Fe) ₃ (Si,Al) ₄ O ₁₀ (OH) ₂ .(Mg,Fe) ₃ (OH) ₆	3.1-3.2
Prehnite	Ca ₂ Al(AlSi ₃ O ₁₀)(OH) ₂	2.8-2.95
Epidote	Ca ₂ (Al,Fe)Al ₂ O(SiO ₄)Si ₂ O ₇ (OH)	3.35-3.45
Phlogopite	KMg ₃ (AlSi ₃ O ₁₀)(OH) ₂	2.86

APPENDIX C

Chromite grain-size measurements.

Table 1 Measured chromite grain-size distributions for 15 polished sections of crushed sample A1, expressed as area percentage in 13 size classes.

Diameter	1	2	3	4	5	6	7	8	9	10	11	12	13	14	15
0-48 μm	2	2	2	2	2	2	2	2	2	2	2	2	2	2	2
48-101 μm	18	17	15	16	15	17	17	17	17	14	18	17	16	15	18
101-155 μm	27	29	26	26	26	26	27	28	27	27	27	28	26	25	2
155-208 μm	25	23	23	23	22	22	23	22	23	24	21	24	25	22	22
208-262 μm	13	15	16	16	17	16	16	15	15	17	15	16	15	15	16
262-315 μm	11	8	10	10	8	9	9	9	9	9	9	6	9	11	8
315-368 μm	3	3	5	4	5	4	4	5	5	5	4	5	4	4	3
368-422 μm	1	2	3	2	3	2	2	2	2	1	3	1	2	3	1
422-475 μm	0	1	1	0	1	2	1	1	1	1	1	1	1	1	1
475-529 μm	1	0	1	1	1	0	0	1	1	0	0	0	0	0	1
529-582 μm	0	0	0	0	0	0	1	0	1	0	1	0	1	1	0
582-636 μm	0	0	0	0	0	1	0	0	0	0	0	0	0	1	0
>636 μm	0	0	0	0	0	0	0	0	0	0	1	0	0	0	0
Median (μm)	164	160	173	170	171	168	166	165	166	171	165	163	170	174	160

Average measured median chromite size: **167 μm**

Using resampling statistics it was determined that measurements on five polished sections ($\pm 15\ 000$ chromite grains) would provide an average value for the median chromite grain-size between 163 and 170 μm , i.e. a relative error of $\sim 2\%$ (95% confidence level).

Table 2 Measured chromite grain-size distributions for 12 traverses on a polished section of milled feed of sample A1, expressed as area percentage in 13 size classes.

Traverse no.	1	2	3	4	5	6	7	8	9	10	11	12
No. grains	3068	2990	2985	2855	2988	2998	2948	2968	2980	2915	2887	2843
<i>0-17 μm</i>	6	6	6	6	6	5	6	6	6	6	6	6
<i>17-36 μm</i>	25	26	25	25	23	25	25	24	25	27	27	26
<i>36-55 μm</i>	26	24	25	27	28	25	29	27	27	27	25	24
<i>55-74 μm</i>	22	20	21	21	22	21	21	23	22	20	20	23
<i>74-94 μm</i>	13	15	15	12	14	14	13	15	15	13	12	16
<i>94-113 μm</i>	6	6	7	6	6	8	6	5	5	6	8	4
<i>113-132 μm</i>	3	3	3	3	1	1	1	1	1	3	1	2
<i>132-151 μm</i>	0	0	0	0	0	1	0	0	0	1	1	0
<i>151-170 μm</i>	0	1	0	0	0	0	0	0	0	0	0	0
<i>170-189 μm</i>	0	0	0	0	0	0	0	0	0	0	0	0
<i>189-208 μm</i>	0	0	0	0	0	0	0	0	0	0	0	0
<i>208-227 μm</i>	0	0	0	0	0	0	0	0	0	0	0	0
<i>>227 μm</i>	1	2	3	4	5	6	7	8	9	10	11	12
Median ECD (μm)	50	51	51	50	51	52	49	51	50	49	49	50

Average measured median chromite size: **50 μm**

Using resampling statistics it was determined that measurements along two traverses of a polished section ($\pm 5\ 000$ chromite grains) would provide an average value for the median chromite grain-size between 49 and 51 μm, i.e. a relative error of <2% (95% confidence level).

APPENDIX D

Electron microprobe analyses of selected chromite grains in samples A1, A4, B4 and C1. All analyses were performed in duplicate and are reported as mass % oxide. The number of cations was calculated on the basis of 32 oxygen anions. $Fe^{2+}:Fe^{3+}$ ratios were calculated on the assumption of stoichiometry. Analytical conditions and minimum detection limits are discussed in Chapter 3.

Table 1 Electron-microprobe analyses of selected chromite grains in sample A1.

	1	2	3	4	5	6	7	8	9	10	11	12
TiO₂	0.83	0.84	0.87	0.84	0.86	0.84	0.80	0.84	0.85	0.90	0.89	0.87
V₂O₃	0.25	0.28	0.28	0.33	0.30	0.35	0.30	0.32	0.34	0.31	0.33	0.32
Al₂O₃	16.59	16.61	16.81	16.95	17.00	16.76	16.76	16.60	16.75	17.40	16.63	16.71
Cr₂O₃	42.53	42.47	42.06	42.06	42.13	42.38	42.45	42.68	42.62	41.25	41.94	41.93
Fe₂O₃	9.58	9.56	9.72	9.62	9.47	9.35	9.28	9.25	9.38	9.94	9.94	9.83
FeO	19.29	19.34	19.21	19.16	19.27	19.47	19.56	19.53	19.46	19.40	19.56	19.40
MnO	0.25	0.28	0.26	0.26	0.26	0.26	0.27	0.26	0.26	0.27	0.26	0.25
NiO	0.16	0.16	0.17	0.15	0.15	0.15	0.15	0.15	0.15	0.17	0.16	0.15
CoO	0.00	0.05	0.04	0.01	0.05	0.04	0.01	0.01	0.02	0.00	0.00	0.03
Cu₂O	0.03	0.01	0.02	0.00	0.07	0.01	0.00	0.02	0.01	0.00	0.06	0.03
ZnO	0.10	0.07	0.08	0.11	0.10	0.01	0.07	0.07	0.14	0.11	0.04	0.03
MgO	9.56	9.51	9.62	9.69	9.57	9.49	9.41	9.42	9.50	9.60	9.41	9.51
Total	99.03	99.05	99.00	99.03	99.09	98.96	98.91	99.01	99.32	99.21	99.07	98.93
Number of ions on the basis of 32 O												
Ti	0.16	0.16	0.17	0.16	0.17	0.16	0.16	0.17	0.17	0.18	0.17	0.17
V	0.05	0.06	0.06	0.07	0.06	0.07	0.06	0.07	0.07	0.07	0.07	0.07
Al	5.11	5.12	5.17	5.21	5.22	5.16	5.17	5.12	5.14	5.33	5.13	5.15
Cr	8.79	8.78	8.68	8.67	8.69	8.76	8.78	8.83	8.78	8.48	8.67	8.67
Fe³⁺	1.89	1.88	1.91	1.89	1.86	1.84	1.83	1.82	1.84	1.94	1.96	1.94
Fe²⁺	4.22	4.23	4.20	4.18	4.20	4.26	4.28	4.27	4.24	4.22	4.28	4.24
Mn	0.06	0.06	0.06	0.06	0.06	0.06	0.06	0.06	0.06	0.06	0.06	0.06
Ni	0.03	0.03	0.04	0.03	0.03	0.03	0.03	0.03	0.03	0.04	0.03	0.03
Co	0.00	0.01	0.01	0.00	0.01	0.01	0.00	0.00	0.00	0.00	0.00	0.01
Cu	0.01	0.00	0.00	0.00	0.02	0.00	0.00	0.00	0.00	0.00	0.01	0.01
Zn	0.02	0.01	0.01	0.02	0.02	0.00	0.01	0.01	0.03	0.02	0.01	0.01
Mg	3.73	3.71	3.74	3.77	3.72	3.70	3.67	3.68	3.69	3.72	3.67	3.71
Σ(M³⁺,M⁴⁺)	16.00	16.00	16.00	16.00	16.00	16.00	16.00	16.00	16.00	16.00	16.00	16.00
ΣM²⁺	8.06	8.06	8.06	8.06	8.06	8.06	8.06	8.06	8.05	8.06	8.06	8.06

Table 1 continued Electron-microprobe analyses of selected chromite grains in sample A1.

	13	14	15	16	17	18	19	20	21	22	23	24
TiO₂	0.87	0.87	0.99	0.98	0.97	0.98	0.97	0.96	0.96	0.97	0.93	0.89
V₂O₃	0.30	0.35	0.38	0.46	0.39	0.41	0.44	0.37	0.45	0.43	0.34	0.42
Al₂O₃	16.65	16.59	15.31	15.45	15.27	15.39	15.44	15.35	15.25	15.33	15.35	15.41
Cr₂O₃	41.81	42.42	44.49	44.18	44.45	44.48	44.53	44.41	44.43	44.45	44.21	44.57
Fe₂O₃	10.01	9.57	8.71	8.73	8.70	8.56	8.54	8.61	8.79	8.74	8.69	8.45
FeO	19.32	19.45	19.74	19.83	19.72	19.87	19.66	19.83	19.78	19.73	19.88	19.87
MnO	0.26	0.26	0.27	0.26	0.28	0.27	0.26	0.28	0.27	0.27	0.26	0.26
NiO	0.15	0.15	0.12	0.10	0.13	0.13	0.10	0.12	0.10	0.12	0.13	0.12
CoO	0.05	0.00	0.03	0.01	0.06	0.01	0.00	0.03	0.00	0.02	0.02	0.02
Cu₂O	0.00	0.00	0.00	0.03	0.03	0.01	0.12	0.00	0.00	0.02	0.02	0.01
ZnO	0.04	0.05	0.03	0.07	0.09	0.07	0.07	0.06	0.11	0.09	0.08	0.12
MgO	9.53	9.52	9.23	9.16	9.14	9.12	9.23	9.11	9.16	9.20	9.04	9.09
Total	98.85	99.09	99.14	99.13	99.07	99.15	99.23	98.99	99.16	99.24	98.79	99.09
Number of ions on the basis of 32 O												
Ti	0.17	0.17	0.20	0.19	0.19	0.19	0.19	0.19	0.19	0.19	0.18	0.18
V	0.06	0.07	0.08	0.10	0.08	0.09	0.09	0.08	0.09	0.09	0.07	0.09
Al	5.14	5.11	4.75	4.79	4.74	4.77	4.78	4.77	4.73	4.75	4.78	4.78
Cr	8.66	8.76	9.25	9.19	9.26	9.25	9.25	9.25	9.24	9.24	9.24	9.28
Fe³⁺	1.97	1.88	1.72	1.73	1.72	1.69	1.69	1.71	1.74	1.73	1.73	1.67
Fe²⁺	4.23	4.25	4.34	4.36	4.34	4.37	4.32	4.37	4.35	4.34	4.39	4.37
Mn	0.06	0.06	0.06	0.06	0.06	0.06	0.06	0.06	0.06	0.06	0.06	0.06
Ni	0.03	0.03	0.03	0.02	0.03	0.03	0.02	0.03	0.02	0.03	0.03	0.02
Co	0.01	0.00	0.01	0.00	0.01	0.00	0.00	0.01	0.00	0.00	0.00	0.00
Cu	0.00	0.00	0.00	0.01	0.01	0.00	0.03	0.00	0.00	0.01	0.00	0.00
Zn	0.01	0.01	0.01	0.01	0.02	0.01	0.01	0.01	0.02	0.02	0.02	0.02
Mg	3.72	3.71	3.62	3.59	3.59	3.58	3.61	3.58	3.60	3.60	3.56	3.57
Σ(M³⁺,M⁴⁺)	16.00	16.00	16.00	16.00	16.00	16.00	16.00	16.00	16.00	16.00	16.00	16.00
ΣM²⁺	8.06	8.06	8.06	8.06	8.06	8.06	8.05	8.06	8.06	8.05	8.06	8.05

Table 1 continued Electron-microprobe analyses of selected chromite grains in sample A1.

	25	26	27	28	29	30	31	32	33	34	35	36
TiO₂	0.97	0.96	0.73	0.76	0.76	0.77	0.78	0.77	0.75	0.77	0.74	0.73
V₂O₃	0.36	0.44	0.32	0.34	0.28	0.29	0.24	0.22	0.32	0.23	0.26	0.30
Al₂O₃	15.35	15.38	17.64	17.82	18.03	17.98	18.26	18.11	18.41	18.32	17.99	18.35
Cr₂O₃	44.18	44.21	42.54	42.30	42.34	42.24	41.84	41.94	41.75	41.59	42.24	41.99
Fe₂O₃	8.77	8.70	8.56	8.59	8.41	8.52	8.64	8.49	8.56	8.80	8.59	8.54
FeO	19.91	19.76	18.71	18.64	18.65	18.62	18.60	18.69	18.56	18.47	18.55	18.39
MnO	0.27	0.28	0.26	0.26	0.26	0.26	0.25	0.25	0.25	0.25	0.25	0.25
NiO	0.12	0.13	0.16	0.15	0.15	0.14	0.15	0.14	0.14	0.16	0.15	0.14
CoO	0.00	0.04	0.00	0.02	0.02	0.03	0.03	0.01	0.02	0.01	0.02	0.04
Cu₂O	0.01	0.03	0.06	0.01	0.02	0.01	0.00	0.02	0.03	0.01	0.01	0.00
ZnO	0.08	0.11	0.09	0.09	0.03	0.08	0.07	0.06	0.12	0.06	0.11	0.15
MgO	9.06	9.10	10.01	10.10	10.15	10.13	10.17	10.06	10.20	10.26	10.17	10.30
Total	98.92	98.99	98.96	98.93	98.99	98.95	98.92	98.65	98.98	98.80	98.96	99.05
Number of ions on the basis of 32 O												
Ti	0.19	0.19	0.14	0.15	0.15	0.15	0.15	0.15	0.15	0.15	0.14	0.14
V	0.08	0.09	0.07	0.07	0.06	0.06	0.05	0.05	0.07	0.05	0.05	0.06
Al	4.77	4.78	5.39	5.44	5.50	5.49	5.56	5.54	5.60	5.58	5.49	5.58
Cr	9.22	9.21	8.73	8.67	8.66	8.64	8.55	8.61	8.52	8.51	8.64	8.56
Fe³⁺	1.74	1.73	1.67	1.67	1.64	1.66	1.68	1.66	1.66	1.71	1.67	1.66
Fe²⁺	4.39	4.36	4.06	4.04	4.03	4.03	4.02	4.06	4.01	3.99	4.02	3.97
Mn	0.06	0.06	0.06	0.06	0.06	0.06	0.06	0.06	0.06	0.06	0.05	0.06
Ni	0.03	0.03	0.03	0.03	0.03	0.03	0.03	0.03	0.03	0.03	0.03	0.03
Co	0.00	0.01	0.00	0.00	0.00	0.01	0.01	0.00	0.00	0.00	0.00	0.01
Cu	0.00	0.01	0.01	0.00	0.01	0.00	0.00	0.01	0.01	0.00	0.00	0.00
Zn	0.02	0.02	0.02	0.02	0.01	0.02	0.01	0.01	0.02	0.01	0.02	0.03
Mg	3.56	3.58	3.87	3.90	3.91	3.91	3.92	3.89	3.93	3.96	3.92	3.96
Σ(M³⁺,M⁴⁺)	16.00	16.00	16.00	16.00	16.00	16.00	16.00	16.00	16.00	16.00	16.00	16.00
ΣM²⁺	8.06	8.06	8.05	8.05	8.05	8.05	8.05	8.06	8.05	8.05	8.05	8.05

Table 1 continued Electron-microprobe analyses of selected chromite grains in sample A1.

	37	38	39	40	41	\bar{x}	s	CL
TiO₂	0.76	0.78	0.75	0.79	0.74	0.85	0.09	0.03
V₂O₃	0.30	0.25	0.31	0.28	0.27	0.33	0.06	0.02
Al₂O₃	17.95	18.28	18.23	18.23	18.14	16.85	1.14	0.36
Cr₂O₃	42.29	41.97	42.11	42.04	41.85	42.79	1.08	0.34
Fe₂O₃	8.60	8.56	8.49	8.47	8.69	8.95	0.51	0.16
FeO	18.49	18.45	18.57	18.51	18.47	19.20	0.53	0.17
MnO	0.25	0.26	0.23	0.25	0.26	0.26	0.01	0.00
NiO	0.16	0.14	0.14	0.16	0.15	0.14	0.02	0.01
CoO	0.00	0.01	0.06	0.00	0.00	0.02	0.02	0.01
Cu₂O	0.05	0.02	0.00	0.05	0.07	0.02	0.03	0.01
ZnO	0.13	0.14	0.10	0.09	0.11	0.08	0.03	0.01
MgO	10.18	10.25	10.21	10.21	10.17	9.65	0.44	0.14
CaO	0.02	0.00	0.00	0.01	0.00	0.00	0.00	0.00
Total	99.04	99.01	99.09	98.97	98.81	99.01	0.13	0.04
Number of ions on the basis of 32 O								
Ti	0.15	0.15	0.15	0.15	0.14	0.17	0.02	0.01
V	0.06	0.05	0.06	0.06	0.06	0.07	0.01	0.00
Al	5.47	5.56	5.55	5.55	5.54	5.18	0.31	0.10
Cr	8.65	8.57	8.59	8.59	8.57	8.83	0.28	0.09
Fe³⁺	1.67	1.66	1.65	1.65	1.69	1.76	0.10	0.03
Fe²⁺	4.00	3.98	4.01	4.00	4.00	4.19	0.15	0.05
Mn	0.05	0.06	0.05	0.06	0.06	0.06	0.00	0.00
Ni	0.03	0.03	0.03	0.03	0.03	0.03	0.00	0.00
Co	0.00	0.00	0.01	0.00	0.00	0.00	0.00	0.00
Cu	0.01	0.01	0.00	0.01	0.02	0.00	0.01	0.00
Zn	0.03	0.03	0.02	0.02	0.02	0.02	0.01	0.00
Mg	3.92	3.95	3.93	3.93	3.93	3.75	0.14	0.04
Σ(M³⁺,M⁴⁺)	16.00	16.00	16.00	16.00	16.00	16.00	0.00	0.00
ΣM²⁺	8.05	8.05	8.05	8.05	8.05	8.05		

x = Average *s* = standard deviation *CL* = 95% confidence limit

Table 2 Electron-microprobe analyses of selected sintered chromite grains in sample A1.

	1	2	3	4	5	6	7	8	\bar{x}	s	CL
TiO₂	0.94	0.99	0.94	1.19	1.13	1.15	1.04	1.15	1.07	0.10	0.08
V₂O₃	0.19	0.29	0.32	0.39	0.36	0.33	0.43	0.46	0.35	0.09	0.07
Al₂O₃	15.69	15.07	16.56	14.77	14.74	14.73	14.60	14.23	15.05	0.74	0.62
Cr₂O₃	43.40	44.26	43.16	44.14	43.82	43.87	44.44	44.58	43.96	0.50	0.42
Fe₂O₃	9.46	9.30	9.17	9.41	9.67	9.69	9.26	9.42	9.42	0.19	0.16
FeO	19.66	19.73	18.41	19.74	19.71	19.60	20.15	19.80	19.60	0.51	0.43
MnO	0.27	0.26	0.26	0.26	0.27	0.27	0.27	0.27	0.27	0.00	0.00
NiO	0.13	0.15	0.16	0.12	0.11	0.13	0.12	0.11	0.13	0.02	0.02
CoO	0.01	0.01	0.04	0.02	0.01	0.00	0.00	0.01	0.01	0.01	0.01
Cu₂O	0.09	0.05	0.02	0.07	0.00	0.03	0.00	0.01	0.03	0.03	0.03
ZnO	0.01	0.09	0.04	0.05	0.05	0.12	0.10	0.10	0.07	0.04	0.03
MgO	9.23	9.14	10.16	9.12	9.14	9.16	8.85	9.02	9.23	0.39	0.33
Total	98.93	99.19	99.08	99.11	98.84	98.89	99.10	99.00	99.02	0.12	0.10
Number of ions on the basis of 32 O											
Ti	0.19	0.20	0.18	0.24	0.22	0.23	0.21	0.23	0.21	0.02	0.02
V	0.04	0.06	0.07	0.08	0.08	0.07	0.09	0.10	0.07	0.02	0.02
Al	4.87	4.68	5.08	4.60	4.60	4.59	4.55	4.45	4.68	0.20	0.17
Cr	9.03	9.22	8.88	9.22	9.17	9.18	9.30	9.35	9.17	0.15	0.13
Fe³⁺	1.87	1.84	1.80	1.87	1.93	1.93	1.85	1.88	1.87	0.04	0.04
Fe²⁺	4.33	4.35	4.01	4.36	4.36	4.34	4.46	4.39	4.32	0.14	0.11
Mn	0.06	0.06	0.06	0.06	0.06	0.06	0.06	0.06	0.06	0.00	0.00
Ni	0.03	0.03	0.03	0.03	0.02	0.03	0.03	0.02	0.03	0.00	0.00
Co	0.00	0.00	0.01	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00
Cu	0.02	0.01	0.00	0.02	0.00	0.01	0.00	0.00	0.01	0.01	0.01
Zn	0.00	0.02	0.01	0.01	0.01	0.02	0.02	0.02	0.01	0.01	0.01
Mg	3.62	3.59	3.94	3.59	3.61	3.61	3.49	3.56	3.63	0.13	0.11
Σ(M³⁺,M⁴⁺)	16.00	16.00	16.00	16.00	16.00	16.00	16.00	16.00	16.00		
ΣM²⁺	8.06	8.06	8.06	8.06	8.07	8.07	8.06	8.06	8.06		

\bar{x} = Average s = standard deviation CL = 95% confidence limit

Table 3 continued Electron-microprobe analyses of selected chromite grains in sample C1.

	13	14	15	16	17	18	19	20	21	22	\bar{x}	s	CL
TiO₂	1.04	1.03	1.04	1.04	1.03	1.03	1.05	1.03	1.05	1.05	0.91	0.21	0.15
V₂O₃	0.36	0.38	0.41	0.40	0.46	0.47	0.42	0.47	0.38	0.33	0.45	0.07	0.05
Al₂O₃	15.49	15.43	15.50	15.43	15.44	15.51	15.56	15.42	15.58	15.55	14.96	0.61	0.44
Cr₂O₃	43.30	43.15	43.14	43.44	43.31	43.32	43.33	43.52	43.28	43.27	43.92	0.88	0.63
Fe₂O₃	9.81	9.84	9.77	9.58	9.72	9.67	9.60	9.58	9.66	9.69	9.44	0.48	0.34
FeO	19.88	19.96	19.89	20.04	19.97	20.03	20.00	20.02	20.01	20.03	20.58	0.77	0.55
MnO	0.27	0.27	0.26	0.26	0.28	0.28	0.26	0.27	0.26	0.25	0.28	0.01	0.01
NiO	0.16	0.14	0.15	0.15	0.16	0.15	0.15	0.16	0.15	0.16	0.13	0.03	0.02
CoO	0.01	0.00	0.01	0.00	0.03	0.00	0.04	0.03	0.02	0.02	0.02	0.02	0.01
Cu₂O	0.03	0.00	0.01	0.01	0.02	0.03	0.00	0.00	0.01	0.00	0.01	0.02	0.01
ZnO	0.10	0.08	0.11	0.12	0.08	0.07	0.11	0.13	0.09	0.09	0.11	0.02	0.02
MgO	9.12	9.07	9.11	9.02	9.06	9.06	9.06	9.03	9.08	9.05	8.59	0.59	0.42
Total	99.41	99.20	99.23	99.32	99.40	99.47	99.42	99.51	99.40	99.32	99.25		
Number of ions on the basis of 32 O													
Ti	0.21	0.20	0.21	0.21	0.20	0.20	0.21	0.20	0.21	0.21	0.18	0.04	0.03
V	0.08	0.08	0.09	0.09	0.10	0.10	0.09	0.10	0.08	0.07	0.10	0.02	0.01
Al	4.79	4.79	4.80	4.78	4.78	4.80	4.81	4.77	4.82	4.82	4.66	0.16	0.11
Cr	8.99	8.98	8.97	9.03	9.00	8.99	8.99	9.03	8.98	8.99	9.18	0.25	0.18
Fe³⁺	1.94	1.95	1.93	1.90	1.92	1.91	1.90	1.89	1.91	1.92	1.88	0.09	0.06
Fe²⁺	4.36	4.39	4.37	4.41	4.39	4.40	4.39	4.39	4.39	4.40	4.55	0.20	0.15
Mn	0.06	0.06	0.06	0.06	0.06	0.06	0.06	0.06	0.06	0.05	0.06	0.00	0.00
Ni	0.03	0.03	0.03	0.03	0.03	0.03	0.03	0.03	0.03	0.03	0.03	0.01	0.00
Co	0.00	0.00	0.00	0.00	0.01	0.00	0.01	0.01	0.00	0.00	0.00	0.00	0.00
Cu	0.01	0.00	0.00	0.00	0.00	0.01	0.00	0.00	0.00	0.00	0.00	0.00	0.00
Zn	0.02	0.01	0.02	0.02	0.02	0.01	0.02	0.03	0.02	0.02	0.02	0.00	0.00
Mg	3.57	3.56	3.57	3.54	3.55	3.54	3.55	3.54	3.55	3.54	3.39	0.21	0.15
Σ(M³⁺,M⁴⁺)	16.00	16.00	16.00	16.00	16.00	16.00	16.00	16.00	16.00	16.00	16.00		
ΣM²⁺	8.06	8.06	8.06	8.06	8.06	8.06	8.06	8.06	8.05	8.06	8.06		

\bar{x} = Average s = standard deviation CL = 95% confidence limit

Table 4 Electron-microprobe analyses of selected sintered chromite grains in sample C1.

	1	2	3	4	5	6	7	8	9	10	11	\bar{x}	s	CL
TiO₂	1.19	1.17	1.18	1.19	1.19	3.45	1.19	1.20	1.20	1.18	1.18	1.19	0.01	0.01
V₂O₃	0.43	0.51	0.47	0.40	0.48	0.40	0.42	0.49	0.43	0.36	0.35	0.44	0.05	0.04
Al₂O₃	15.15	15.18	15.20	15.27	15.19	14.54	15.16	15.08	14.89	15.16	15.26	15.16	0.11	0.08
Cr₂O₃	43.32	43.41	43.34	43.42	43.45	42.87	43.38	43.55	43.74	43.78	43.55	43.49	0.16	0.11
Fe₂O₃	9.98	9.89	9.92	9.84	9.91	8.46	9.97	9.79	9.76	9.73	9.96	9.87	0.09	0.07
FeO	19.47	19.48	19.51	19.56	19.48	20.67	19.42	19.62	19.68	19.54	19.42	19.52	0.08	0.06
MnO	0.28	0.28	0.28	0.30	0.29	0.25	0.27	0.28	0.29	0.28	0.27	0.28	0.01	0.01
NiO	0.17	0.15	0.14	0.16	0.17	0.14	0.15	0.14	0.16	0.16	0.17	0.16	0.01	0.01
CoO	0.00	0.04	0.04	0.00	0.04	0.03	0.04	0.02	0.02	0.00	0.03	0.02	0.02	0.01
Cu₂O	0.00	0.00	0.01	0.00	0.03	0.04	0.02	0.03	0.04	0.05	0.00	0.02	0.02	0.01
ZnO	0.05	0.09	0.10	0.10	0.12	0.11	0.10	0.11	0.11	0.10	0.13	0.10	0.02	0.02
MgO	9.38	9.37	9.33	9.32	9.33	8.61	9.38	9.25	9.15	9.31	9.42	9.32	0.08	0.05
Total	99.23	99.38	99.35	99.37	99.49	99.12	99.32	99.38	99.29	99.49	99.56	99.39	0.10	0.07
Number of ions on the basis of 32 O														
Ti	0.23	0.23	0.23	0.23	0.23	0.69	0.24	0.24	0.24	0.23	0.23	0.23	0.00	0.00
V	0.09	0.11	0.10	0.08	0.10	0.09	0.09	0.10	0.09	0.08	0.07	0.09	0.01	0.01
Al	4.69	4.70	4.71	4.73	4.70	4.55	4.69	4.67	4.62	4.69	4.71	4.69	0.03	0.02
Cr	9.00	9.01	9.00	9.01	9.01	8.99	9.01	9.05	9.11	9.08	9.02	9.03	0.04	0.03
Fe³⁺	1.98	1.95	1.96	1.94	1.96	1.69	1.97	1.94	1.93	1.92	1.96	1.95	0.02	0.01
Fe²⁺	4.28	4.28	4.29	4.29	4.27	4.59	4.27	4.31	4.34	4.29	4.25	4.29	0.02	0.02
Mn	0.06	0.06	0.06	0.07	0.06	0.06	0.06	0.06	0.06	0.06	0.06	0.06	0.00	0.00
Ni	0.04	0.03	0.03	0.03	0.04	0.03	0.03	0.03	0.03	0.03	0.04	0.03	0.00	0.00
Co	0.00	0.01	0.01	0.00	0.01	0.01	0.01	0.00	0.00	0.00	0.01	0.00	0.00	0.00
Cu	0.00	0.00	0.00	0.00	0.01	0.01	0.00	0.01	0.01	0.01	0.00	0.00	0.00	0.00
Zn	0.01	0.02	0.02	0.02	0.02	0.02	0.02	0.02	0.02	0.02	0.02	0.02	0.00	0.00
Mg	3.68	3.67	3.65	3.65	3.65	3.41	3.67	3.63	3.59	3.64	3.68	3.65	0.03	0.02
Σ(M³⁺,M⁴⁺)	16.00	16.00	16.00	16.00	16.00	16.00	16.00	16.00	16.00	16.00	16.00	16.00		
ΣM²⁺	8.06	8.06	8.06	8.06	8.06	8.11	8.06	8.06	8.06	8.06	8.06	8.06		

\bar{x} = Average s = standard deviation CL = 95% confidence limit

Table 5 Electron-microprobe analyses of selected sintered chromite grains in sample C1.

	1	2	3	4	5	6	7	8	9	10	\bar{x}	s	CL
TiO₂	1.31	1.22	1.30	1.31	1.30	1.29	1.30	1.30	1.31	1.30	1.29	0.03	0.02
V₂O₃	0.46	0.31	0.39	0.35	0.44	0.36	0.42	0.44	0.43	0.43	0.40	0.05	0.03
Al₂O₃	15.00	15.01	15.19	15.27	15.22	15.23	15.19	15.22	15.31	15.21	15.19	0.10	0.07
Cr₂O₃	42.94	42.95	43.07	43.02	42.94	42.80	42.78	42.97	42.94	42.96	42.94	0.09	0.06
Fe₂O₃	10.24	10.44	10.14	10.08	10.09	10.25	10.25	10.01	10.06	10.02	10.16	0.14	0.10
FeO	20.11	20.04	20.05	20.22	20.25	20.11	20.08	20.22	20.10	20.19	20.14	0.08	0.06
MnO	0.27	0.29	0.25	0.24	0.26	0.27	0.25	0.26	0.27	0.26	0.26	0.01	0.01
NiO	0.16	0.15	0.16	0.16	0.17	0.17	0.16	0.16	0.15	0.18	0.16	0.01	0.01
CoO	0.02	0.02	0.02	0.01	0.00	0.01	0.03	0.00	0.03	0.03	0.02	0.01	0.01
Cu₂O	0.03	0.08	0.08	0.05	0.02	0.00	0.03	0.01	0.04	0.01	0.04	0.03	0.02
ZnO	0.09	0.08	0.11	0.11	0.10	0.11	0.10	0.11	0.15	0.14	0.11	0.02	0.02
MgO	8.93	8.93	8.99	8.91	8.90	8.96	8.97	8.91	8.96	8.88	8.93	0.04	0.03
Total	99.37	99.36	99.57	99.55	99.49	99.37	99.37	99.41	99.55	99.43	99.45	0.08	0.06
Number of ions on the basis of 32 O													
Ti	0.26	0.24	0.26	0.26	0.26	0.26	0.26	0.26	0.26	0.26	0.26	0.00	0.00
V	0.10	0.07	0.08	0.07	0.09	0.08	0.09	0.09	0.09	0.09	0.08	0.01	0.01
Al	4.66	4.66	4.71	4.73	4.72	4.73	4.72	4.72	4.74	4.72	4.71	0.03	0.02
Cr	8.95	8.95	8.95	8.94	8.93	8.91	8.91	8.94	8.92	8.94	8.94	0.02	0.01
Fe³⁺	2.03	2.07	2.01	1.99	2.00	2.03	2.03	1.98	1.99	1.99	2.01	0.03	0.02
Fe²⁺	4.43	4.42	4.41	4.45	4.46	4.43	4.42	4.45	4.42	4.45	4.43	0.02	0.01
Mn	0.06	0.06	0.05	0.05	0.06	0.06	0.06	0.06	0.06	0.06	0.06	0.00	0.00
Ni	0.03	0.03	0.03	0.03	0.04	0.04	0.03	0.03	0.03	0.04	0.03	0.00	0.00
Co	0.00	0.01	0.00	0.00	0.00	0.00	0.01	0.00	0.01	0.01	0.00	0.00	0.00
Cu	0.01	0.02	0.02	0.01	0.00	0.00	0.01	0.00	0.01	0.00	0.01	0.01	0.00
Zn	0.02	0.02	0.02	0.02	0.02	0.02	0.02	0.02	0.03	0.03	0.02	0.00	0.00
Mg	3.51	3.51	3.52	3.49	3.49	3.52	3.52	3.50	3.51	3.48	3.50	0.01	0.01
Σ(M³⁺,M⁴⁺)	16.00	16.00	16.00	16.00	16.00	16.00	16.00	16.00	16.00	16.00	16.00		
ΣM²⁺	8.06	8.06	8.06	8.06	8.06	8.06	8.06	8.06	8.06	8.06	8.06		

\bar{x} = Average s = standard deviation CL = 95% confidence limit

Table 6 Electron-microprobe analyses of selected chromite grains in sample A5.

	1	2	3	4	5	6	7	8	9	10	11	12	13
TiO₂	1.04	1.04	1.04	1.04	1.04	1.08	1.06	1.03	1.05	1.01	1.07	0.93	0.96
V₂O₃	0.29	0.23	0.23	0.26	0.32	0.27	0.24	0.27	0.25	0.26	0.23	0.22	0.27
Al₂O₃	17.64	17.51	17.23	17.42	17.31	17.78	17.11	16.99	17.25	16.92	17.15	17.38	17.11
Cr₂O₃	44.45	44.57	44.78	44.32	44.60	44.07	44.65	44.96	44.77	44.72	44.56	44.50	44.18
Fe₂O₃	5.70	5.76	5.86	5.98	5.76	5.93	5.94	5.83	5.90	6.01	6.04	5.99	6.30
FeO	21.96	21.98	21.98	21.88	22.06	21.76	22.01	22.05	21.93	21.74	21.92	22.21	22.28
MnO	0.27	0.26	0.25	0.28	0.25	0.26	0.26	0.27	0.26	0.26	0.25	0.28	0.26
NiO	0.19	0.17	0.15	0.17	0.17	0.17	0.16	0.16	0.17	0.19	0.16	0.17	0.17
CoO	0.01	0.03	0.01	0.02	0.00	0.02	0.01	0.02	0.02	0.02	0.03	0.00	0.02
Cu₂O	0.00	0.04	0.01	0.00	0.03	0.00	0.00	0.04	0.03	0.07	0.07	0.01	0.00
ZnO	0.13	0.10	0.08	0.12	0.11	0.12	0.11	0.09	0.09	0.13	0.09	0.12	0.10
MgO	8.01	7.98	8.00	8.01	7.92	8.16	7.92	7.88	8.01	7.96	7.96	7.82	7.71
Total	99.54	99.52	99.48	99.35	99.43	99.46	99.32	99.44	99.59	99.16	99.37	99.48	99.22
Number of ions on the basis of 32 O													
Ti	0.20	0.20	0.20	0.21	0.21	0.21	0.21	0.20	0.21	0.20	0.21	0.18	0.19
V	0.06	0.05	0.05	0.05	0.07	0.06	0.05	0.06	0.05	0.06	0.05	0.05	0.06
Al	5.43	5.40	5.32	5.38	5.35	5.47	5.30	5.26	5.32	5.25	5.30	5.37	5.31
Cr	9.18	9.22	9.27	9.18	9.24	9.10	9.27	9.33	9.26	9.31	9.24	9.22	9.20
Fe³⁺	1.12	1.13	1.16	1.18	1.14	1.16	1.17	1.15	1.16	1.19	1.19	1.18	1.25
Fe²⁺	4.80	4.81	4.81	4.79	4.83	4.75	4.83	4.84	4.80	4.79	4.81	4.87	4.91
Mn	0.06	0.06	0.06	0.06	0.06	0.06	0.06	0.06	0.06	0.06	0.06	0.06	0.06
Ni	0.04	0.04	0.03	0.04	0.04	0.04	0.03	0.03	0.04	0.04	0.03	0.04	0.04
Co	0.00	0.01	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.01	0.00	0.01
Cu	0.00	0.01	0.00	0.00	0.01	0.00	0.00	0.01	0.01	0.01	0.02	0.00	0.00
Zn	0.02	0.02	0.02	0.02	0.02	0.02	0.02	0.02	0.02	0.02	0.02	0.02	0.02
Mg	3.12	3.11	3.12	3.13	3.09	3.17	3.10	3.08	3.12	3.12	3.11	3.05	3.03
Σ(M³⁺,M⁴⁺)	16.00	16.00	16.00	16.00	16.00	16.00	16.00	16.00	16.00	16.00	16.00	16.00	16.00
ΣM²⁺	8.05	8.05	8.05	8.05	8.05	8.05	8.05	8.05	8.05	8.05	8.05	8.04	8.05

Table 6 continued Electron-microprobe analyses of selected chromite grains in sample A5.

	27	28	29	30	31	32	33	34	35	36	37	\bar{x}	s	CL
TiO₂	0.90	0.91	0.92	0.93	0.91	0.94	0.95	0.97	0.95	0.97	0.97	0.99	0.05	0.02
V₂O₃	0.34	0.29	0.27	0.27	0.24	0.28	0.38	0.36	0.33	0.34	0.30	0.27	0.04	0.01
Al₂O₃	18.69	18.31	18.18	17.78	18.11	18.30	17.55	17.56	17.53	17.44	17.38	17.44	0.41	0.14
Cr₂O₃	42.91	43.35	43.35	43.53	43.00	43.03	43.73	43.72	43.73	43.64	43.73	44.18	0.63	0.21
Fe₂O₃	5.97	6.03	6.17	6.28	6.38	6.22	6.28	6.31	6.34	6.48	6.35	6.08	0.27	0.09
FeO	21.58	21.71	21.75	21.75	21.61	21.72	21.86	21.93	21.83	21.95	21.94	21.88	0.28	0.09
MnO	0.25	0.25	0.25	0.27	0.26	0.25	0.26	0.27	0.26	0.25	0.25	0.26	0.01	0.00
NiO	0.16	0.18	0.17	0.17	0.17	0.17	0.16	0.18	0.18	0.18	0.16	0.17	0.01	0.00
CoO	0.01	0.05	0.03	0.00	0.05	0.00	0.04	0.00	0.00	0.02	0.00	0.02	0.02	0.01
Cu₂O	0.06	0.04	0.01	0.02	0.03	0.02	0.01	0.08	0.02	0.00	0.03	0.03	0.02	0.01
ZnO	0.09	0.08	0.09	0.08	0.05	0.09	0.08	0.11	0.11	0.12	0.03	0.10	0.02	0.01
MgO	8.30	8.18	8.16	8.10	8.18	8.20	8.03	7.95	8.03	7.94	7.96	8.01	0.21	0.07
Total	99.12	99.24	99.23	99.03	98.84	99.06	99.17	99.29	99.17	99.19	98.96	99.28	0.16	0.05
Number of ions on the basis of 32 O														
Ti	0.18	0.18	0.18	0.18	0.18	0.19	0.19	0.19	0.19	0.19	0.19	0.20	0.01	0.00
V	0.07	0.06	0.06	0.06	0.05	0.06	0.08	0.08	0.07	0.07	0.06	0.06	0.01	0.00
Al	5.74	5.63	5.60	5.49	5.60	5.64	5.42	5.43	5.42	5.40	5.39	5.39	0.12	0.04
Cr	8.84	8.95	8.95	9.03	8.91	8.89	9.07	9.06	9.07	9.06	9.10	9.16	0.13	0.04
Fe³⁺	1.17	1.18	1.21	1.24	1.26	1.22	1.24	1.24	1.25	1.28	1.26	1.20	0.06	0.02
Fe²⁺	4.70	4.74	4.75	4.77	4.74	4.75	4.80	4.81	4.79	4.82	4.83	4.80	0.07	0.02
Mn	0.06	0.06	0.06	0.06	0.06	0.05	0.06	0.06	0.06	0.06	0.06	0.06	0.00	0.00
Ni	0.03	0.04	0.04	0.04	0.04	0.03	0.03	0.04	0.04	0.04	0.03	0.04	0.00	0.00
Co	0.00	0.01	0.01	0.00	0.01	0.00	0.01	0.00	0.00	0.00	0.00	0.00	0.00	0.00
Cu	0.01	0.01	0.00	0.00	0.01	0.00	0.00	0.02	0.00	0.00	0.01	0.01	0.01	0.00
Zn	0.02	0.02	0.02	0.02	0.01	0.02	0.01	0.02	0.02	0.02	0.01	0.02	0.00	0.00
Mg	3.22	3.18	3.18	3.17	3.20	3.19	3.14	3.11	3.14	3.11	3.12	3.13	0.07	0.02
Σ(M³⁺, M⁴⁺)	16.00	16.00	16.00	16.00	16.00	16.00	16.00	16.00	16.00	16.00	16.00	16.00		
ΣM²⁺	8.05	8.05	8.05	8.05	8.05	8.05	8.05	8.05	8.05	8.05	8.05	8.05		

\bar{x} = Average s = standard deviation CL = 95% confidence limit

Table 7 Electron-microprobe analyses of selected sintered chromite grains in sample A5.

	1	2	3	4	5	6	7	8	9	10	11
TiO₂	1.99	2.01	1.99	2.01	2.00	1.97	2.00	2.01	1.89	1.94	3.03
V₂O₃	0.26	0.32	0.33	0.26	0.30	0.24	0.29	0.33	0.23	0.25	0.26
Al₂O₃	14.88	15.13	15.07	15.06	15.03	15.46	15.80	15.70	15.43	15.66	12.71
Cr₂O₃	41.15	40.72	40.80	41.08	41.09	40.43	40.05	40.54	41.28	40.74	40.43
Fe₂O₃	9.43	9.50	9.40	9.25	9.36	9.82	9.87	9.59	9.35	9.57	11.23
FeO	25.16	25.16	25.10	25.24	25.19	24.36	24.38	24.48	24.48	24.54	26.21
MnO	0.28	0.30	0.28	0.30	0.28	0.28	0.30	0.29	0.29	0.31	0.31
NiO	0.16	0.16	0.16	0.17	0.17	0.14	0.16	0.13	0.14	0.14	0.14
CoO	0.00	0.02	0.02	0.02	0.01	0.04	0.00	0.01	0.02	0.01	0.03
Cu₂O	0.06	0.03	0.04	0.01	0.07	0.00	0.05	0.01	0.06	0.00	0.06
ZnO	0.14	0.14	0.14	0.15	0.15	0.15	0.13	0.13	0.13	0.12	0.15
MgO	5.52	5.55	5.58	5.49	5.52	6.12	6.15	6.15	6.08	6.10	4.65
Total	98.75	98.74	98.68	98.74	98.89	98.72	98.88	99.07	99.13	99.10	98.80
Number of ions on the basis of 32 O											
Ti	0.41	0.41	0.41	0.41	0.41	0.40	0.41	0.41	0.38	0.39	0.63
V	0.06	0.07	0.07	0.06	0.06	0.05	0.06	0.07	0.05	0.05	0.06
Al	4.77	4.84	4.82	4.82	4.81	4.92	5.01	4.97	4.89	4.96	4.14
Cr	8.84	8.74	8.76	8.82	8.81	8.63	8.52	8.61	8.78	8.66	8.84
Fe³⁺	1.93	1.94	1.92	1.89	1.91	1.99	2.00	1.94	1.89	1.94	2.34
Fe²⁺	5.72	5.71	5.70	5.73	5.71	5.50	5.49	5.50	5.51	5.52	6.06
Mn	0.06	0.07	0.06	0.07	0.07	0.06	0.07	0.07	0.07	0.07	0.07
Ni	0.04	0.03	0.03	0.04	0.04	0.03	0.03	0.03	0.03	0.03	0.03
Co	0.00	0.00	0.00	0.00	0.00	0.01	0.00	0.00	0.00	0.00	0.01
Cu	0.01	0.01	0.01	0.00	0.02	0.00	0.01	0.00	0.01	0.00	0.01
Zn	0.03	0.03	0.03	0.03	0.03	0.03	0.02	0.03	0.03	0.02	0.03
Mg	2.24	2.24	2.26	2.22	2.23	2.46	2.47	2.46	2.44	2.44	1.92
Σ(M³⁺,M⁴⁺)	16.00	16.00	16.00	16.00	16.00	16.00	16.00	16.00	16.00	16.00	16.00
ΣM²⁺	8.10	8.10	8.10	8.10	8.09	8.10	8.09	8.09	8.08	8.09	8.13

Table 7 continued Electron-microprobe analyses of selected sintered chromite grains in sample A5.

	12	13	14	15	\bar{x}	s	CL
TiO₂	3.00	2.93	2.93	2.98	2.31	0.49	0.28
V₂O₃	0.29	0.29	0.27	0.31	0.28	0.03	0.02
Al₂O₃	12.77	12.67	12.81	12.79	14.46	1.28	0.75
Cr₂O₃	40.49	40.75	40.65	40.57	40.72	0.33	0.19
Fe₂O₃	11.16	11.02	11.03	11.10	10.05	0.80	0.47
FeO	26.21	26.23	26.27	26.33	25.29	0.77	0.45
MnO	0.31	0.31	0.31	0.30	0.30	0.01	0.01
NiO	0.13	0.16	0.14	0.15	0.15	0.01	0.01
CoO	0.07	0.00	0.02	0.02	0.02	0.02	0.01
Cu₂O	0.07	0.04	0.00	0.01	0.03	0.03	0.02
ZnO	0.17	0.16	0.17	0.11	0.14	0.02	0.01
MgO	4.64	4.64	4.66	4.66	5.43	0.63	0.37
Total	98.88	98.81	98.84	98.91	98.86	0.14	0.08
Number of ions on the basis of 32 O							
Ti	0.62	0.61	0.61	0.62	0.47	0.11	0.06
V	0.06	0.07	0.06	0.07	0.06	0.01	0.00
Al	4.16	4.13	4.17	4.16	4.64	0.36	0.21
Cr	8.84	8.90	8.87	8.85	8.77	0.11	0.06
Fe³⁺	2.32	2.29	2.29	2.30	2.06	0.18	0.11
Fe²⁺	6.05	6.06	6.07	6.08	5.76	0.24	0.14
Mn	0.07	0.07	0.07	0.07	0.07	0.00	0.00
Ni	0.03	0.03	0.03	0.03	0.03	0.00	0.00
Co	0.01	0.00	0.00	0.00	0.00	0.00	0.00
Cu	0.02	0.01	0.00	0.00	0.01	0.01	0.00
Zn	0.03	0.03	0.03	0.02	0.03	0.00	0.00
Mg	1.91	1.91	1.92	1.92	2.20	0.23	0.13
Σ(M³⁺,M⁴⁺)	16.00	16.00	16.00	16.00	16.00		
ΣM²⁺	8.13	8.13	8.12	8.12	8.10		

\bar{x} = Average s = standard deviation CL = 95% confidence limit

Table 8 continued Electron-microprobe analyses of selected sintered chromite grains in sample B4.

	15	16	17	18	19	20	21	22	23	24	25	26	27	28
TiO₂	1.32	1.18	1.17	1.30	1.30	1.34	1.32	1.33	1.32	1.29	1.33	1.31	1.31	1.22
V₂O₃	0.42	0.40	0.37	0.47	0.37	0.46	0.44	0.45	0.41	0.36	0.34	0.43	0.32	0.44
Al₂O₃	15.17	14.98	15.42	15.01	14.91	14.81	14.85	14.98	14.87	15.11	14.99	15.08	15.08	15.21
Cr₂O₃	41.58	41.79	41.65	42.05	42.03	41.98	42.18	41.96	41.93	41.48	41.83	41.76	41.72	41.47
Fe₂O₃	10.93	10.97	10.61	10.73	10.88	8.99	8.77	8.88	8.90	9.19	8.84	8.98	8.95	9.06
FeO	20.78	20.94	20.66	20.92	20.88	22.85	23.03	23.01	22.97	22.87	23.06	22.86	23.06	23.00
MnO	0.28	0.28	0.28	0.28	0.26	0.27	0.28	0.29	0.27	0.27	0.28	0.28	0.27	0.26
NiO	0.16	0.15	0.15	0.18	0.16	0.19	0.16	0.19	0.18	0.15	0.16	0.17	0.17	0.16
CoO	0.01	0.01	0.02	0.03	0.01	0.00	0.02	0.00	0.01	0.00	0.02	0.01	0.01	0.00
Cu₂O	0.03	0.02	0.04	0.00	0.00	0.08	0.00	0.04	0.00	0.03	0.00	0.04	0.00	0.02
ZnO	0.04	0.04	0.03	0.02	0.02	0.05	0.05	0.02	0.02	0.04	0.01	0.05	0.03	0.07
MgO	8.47	8.33	8.59	8.41	8.43	8.22	8.20	8.21	8.25	8.25	8.22	8.27	8.21	8.15
Total	98.98	98.92	99.03	99.18	99.05	99.24	99.31	99.34	99.14	99.09	99.08	99.26	99.12	99.06
Number of ions on the basis of 32 O														
Ti	0.26	0.24	0.23	0.26	0.26	0.27	0.26	0.27	0.26	0.26	0.27	0.26	0.26	0.24
V	0.09	0.09	0.08	0.10	0.08	0.10	0.09	0.09	0.09	0.08	0.07	0.09	0.07	0.09
Al	4.74	4.70	4.80	4.69	4.66	4.64	4.65	4.68	4.66	4.73	4.69	4.71	4.72	4.76
Cr	8.72	8.79	8.71	8.81	8.82	8.82	8.85	8.79	8.81	8.71	8.79	8.75	8.76	8.71
Fe³⁺	2.18	2.20	2.11	2.14	2.17	1.80	1.75	1.77	1.78	1.84	1.77	1.79	1.79	1.81
Fe²⁺	4.61	4.66	4.57	4.64	4.64	5.07	5.11	5.10	5.10	5.08	5.12	5.07	5.12	5.11
Mn	0.06	0.06	0.06	0.06	0.06	0.06	0.06	0.07	0.06	0.06	0.06	0.06	0.06	0.06
Ni	0.03	0.03	0.03	0.04	0.03	0.04	0.03	0.04	0.04	0.03	0.03	0.04	0.04	0.03
Co	0.00	0.00	0.00	0.01	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00
Cu	0.01	0.00	0.01	0.00	0.00	0.02	0.00	0.01	0.00	0.01	0.00	0.01	0.00	0.00
Zn	0.01	0.01	0.01	0.00	0.00	0.01	0.01	0.00	0.00	0.01	0.00	0.01	0.01	0.01
Mg	3.35	3.30	3.39	3.32	3.34	3.26	3.24	3.24	3.27	3.26	3.26	3.27	3.25	3.23
Σ(M³⁺,M⁴⁺)	16.00	16.00	16.00	16.00	16.00	15.61	15.60	15.61	15.59	15.61	15.59	15.61	15.60	15.62
ΣM²⁺	8.07	8.07	8.07	8.07	8.07	8.46	8.47	8.46	8.48	8.47	8.48	8.46	8.47	8.45

Table 8 continued Electron-microprobe analyses of selected sintered chromite grains in sample B4.

	29	30	31	32	33	34	35	36	37	38	39	\bar{x}	s	CL
TiO₂	1.29	1.32	1.29	1.29	1.11	1.25	1.14	1.28	1.25	1.18	1.30	1.30	0.08	0.03
V₂O₃	0.43	0.48	0.42	0.38	0.45	0.42	0.48	0.45	0.39	0.36	0.45	0.40	0.05	0.01
Al₂O₃	14.93	15.00	15.10	15.14	15.81	15.32	15.26	15.32	14.83	14.67	14.97	15.01	0.27	0.09
Cr₂O₃	41.87	41.71	41.54	41.65	42.00	41.93	41.54	41.69	42.16	42.33	41.78	41.81	0.24	0.08
Fe₂O₃	8.98	8.93	8.96	9.11	8.32	8.45	8.94	8.69	8.60	8.81	8.80	9.88	1.08	0.35
FeO	22.80	22.99	22.81	22.56	22.44	22.81	22.58	22.84	22.93	22.90	22.89	22.06	0.87	0.28
MnO	0.28	0.27	0.28	0.28	0.28	0.27	0.27	0.28	0.28	0.27	0.28	0.27	0.01	0.00
NiO	0.17	0.17	0.17	0.18	0.15	0.16	0.16	0.16	0.16	0.16	0.16	0.17	0.01	0.00
CoO	0.03	0.02	0.00	0.02	0.01	0.02	0.02	0.04	0.02	0.03	0.04	0.02	0.01	0.00
Cu₂O	0.02	0.00	0.00	0.09	0.04	0.01	0.00	0.04	0.05	0.02	0.02	0.02	0.02	0.01
ZnO	0.11	0.11	0.13	0.10	0.08	0.06	0.10	0.04	0.09	0.11	0.11	0.05	0.04	0.01
MgO	8.24	8.20	8.27	8.35	8.41	8.30	8.24	8.25	8.07	8.01	8.20	8.21	0.16	0.05
Total	99.16	99.20	98.97	99.15	99.12	98.99	98.88	99.09	98.85	98.91	98.99	99.12	0.17	0.06
Number of ions on the basis of 32 O														
Ti	0.26	0.26	0.26	0.26	0.22	0.25	0.23	0.25	0.25	0.24	0.26	0.26	0.02	0.01
V	0.09	0.10	0.09	0.08	0.09	0.09	0.10	0.10	0.08	0.08	0.10	0.09	0.01	0.00
Al	4.67	4.69	4.73	4.73	4.92	4.79	4.78	4.79	4.66	4.61	4.69	4.70	0.07	0.02
Cr	8.79	8.75	8.73	8.73	8.77	8.80	8.73	8.74	8.89	8.93	8.79	8.78	0.06	0.02
Fe³⁺	1.80	1.78	1.79	1.82	1.65	1.69	1.79	1.73	1.73	1.77	1.76	1.97	0.22	0.07
Fe²⁺	5.06	5.10	5.07	5.00	4.96	5.06	5.02	5.07	5.12	5.11	5.09	4.90	0.19	0.06
Mn	0.06	0.06	0.06	0.06	0.06	0.06	0.06	0.06	0.06	0.06	0.06	0.06	0.00	0.00
Ni	0.04	0.04	0.04	0.04	0.03	0.03	0.03	0.03	0.03	0.03	0.03	0.04	0.00	0.00
Co	0.01	0.00	0.00	0.00	0.00	0.00	0.00	0.01	0.00	0.01	0.01	0.00	0.00	0.00
Cu	0.00	0.00	0.00	0.02	0.01	0.00	0.00	0.01	0.01	0.00	0.00	0.00	0.00	0.00
Zn	0.02	0.02	0.02	0.02	0.02	0.01	0.02	0.01	0.02	0.02	0.02	0.01	0.01	0.00
Mg	3.26	3.25	3.28	3.30	3.31	3.28	3.27	3.26	3.21	3.19	3.25	3.25	0.06	0.02
Σ(M³⁺,M⁴⁺)	15.61	15.60	15.60	15.62	15.66	15.61	15.63	15.62	15.62	15.63	15.60	15.80		
ΣM²⁺	8.46	8.47	8.48	8.45	8.40	8.46	8.44	8.45	8.46	8.44	8.47	8.27		

\bar{x} = Average s = standard deviation CL = 95% confidence limit

APPENDIX E

SEM-EDS analyses of selected silicate grains reported as mass % oxide.

- Table 1** Plagioclase analyses: sample A1
- Table 2** Plagioclase analyses: sample B1
- Table 3** Plagioclase analyses: sample B2.
- Table 4** Plagioclase analyses: sample B3.
- Table 5** Plagioclase analyses: sample A4
- Table 6** Plagioclase analyses: sample A5.
- Table 7** Plagioclase analyses: sample C1.
- Table 8** Plagioclase analyses: sample C5.
- Table 9** Pyroxene analyses: sample A1.
- Table 10** Pyroxene analyses: sample B1.
- Table 11** Pyroxene analyses: sample B2 (1-4) and B3 (5-9).
- Table 12** Pyroxene analyses: sample A4.
- Table 13** Pyroxene analyses: sample C1(1-7) and C5 (8-9).
- Table 14** Selected phlogopite analyses.
- Table 15** Selected amphibole analyses: Edenitic hornblende (1,3,5-9) & tremolite (2,4)
- Table 16** Selected pumpellyite analyses.
- Table 17** Selected epidote (1-3) and prehnite (4) analyses.
- Table 18** Selected chlorite (1-2),septechlorite (3-5), serpentine (6-8) and talc (9-10) analyses.

Table 1 Plagioclase analyses: sample A1.

	1	2	3	4	5	6	7	8	9	10	11	12	13	14	15	16	17
Na ₂ O	3.0	3.4	3.0	3.0	3.2	3.2	3.5	3.0	3.1	3.4	3.0	3.3	3.5	4.6	3.1	4.0	4.4
MgO	0.1	0.1	0.1	0.1	0.1	0.1	0.1	0.0	0.1	0.1	0.1	0.1	0.1	0.1	0.1	0.1	0.1
Al ₂ O ₃	32.6	32.2	32.8	32.8	32.4	32.6	31.9	32.9	32.6	32.6	32.5	32.3	31.6	30.4	32.2	31.2	30.8
SiO ₂	49.2	50.0	49.1	49.4	49.6	49.7	50.5	49.5	49.4	50.0	49.1	49.7	50.2	52.9	49.2	51.4	52.5
K ₂ O	0.1	0.1	0.1	0.1	0.1	0.1	0.2	0.1	0.1	0.1	0.1	0.1	0.1	0.1	0.1	0.1	0.1
CaO	15.3	14.7	15.4	15.3	15.1	15.2	14.5	15.4	15.3	14.6	15.2	14.9	14.3	12.4	15.0	13.7	12.9
TiO ₂	0.0	0.1	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.1	0.0	0.0	0.0	0.1	0.0	0.0	0.0
Cr ₂ O ₃	0.2	0.2	0.2	0.3	0.2	0.2	0.1	0.2	0.3	0.1	0.1	0.3	0.2	0.3	0.1	0.2	0.2
MnO	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0
FeO	0.2	0.2	0.2	0.2	0.2	0.2	0.2	0.2	0.2	0.1	0.1	0.2	0.2	0.2	0.1	0.2	0.2
Total	100.7	100.8	100.9	101.2	100.8	101.3	101.0	101.4	101.1	101.1	100.2	100.8	100.2	101.2	100.0	100.9	101.2
Number of ions on the basis of 32 O																	
Na	1.2	1.3	1.2	1.2	1.3	1.3	1.4	1.2	1.2	1.3	1.2	1.3	1.3	1.8	1.2	1.6	1.7
Mg	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0
Al	7.0	6.9	7.0	7.0	6.9	6.9	6.8	7.0	7.0	6.9	7.0	6.9	6.9	6.4	6.9	6.6	6.5
Si	8.9	9.1	8.9	8.9	9.0	9.0	9.1	8.9	8.9	9.0	9.0	9.0	9.0	9.5	9.0	9.3	9.4
K	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0
Ca	3.0	2.9	3.0	3.0	2.9	2.9	2.8	3.0	3.0	2.8	3.0	2.9	2.9	2.4	2.9	2.6	2.5
Ti	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0
Cr	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0
Mn	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0
Fe ²⁺	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0
Z	16.0	16.0	16.0	16.0	16.0	16.0	16.0	16.0	16.0	16.0	16.0	16.0	16.0	16.0	16.0	16.0	16.0
X	4.2	4.2	4.2	4.2	4.2	4.2	4.2	4.2	4.2	4.2	4.2	4.2	4.2	4.2	4.2	4.2	4.2
An	71	68	71	71	69	69	66	71	71	68	71	69	69	56	70	62	59
Or	1	0	1	1	1	1	1	1	1	1	0	1	1	0	1	0	1
Ab	29	32	28	29	30	30	33	28	29	32	29	30	30	43	30	37	40

$$Z = \Sigma(\text{Si}, \text{Al}), X = \Sigma(\text{Na}, \text{Ca}, \text{K})$$

Table 2 Plagioclase analyses: sample B1.

	1	2	3	4	5	6	7	8	9	10	11	12	13	14	15	16
Na ₂ O	3.6	3.5	3.6	3.4	3.8	3.6	3.7	4.1	3.7	4.1	3.5	3.6	3.2	3.7	4.2	4.1
MgO	0.1	0.2	0.1	0.1	0.1	0.1	0.1	0.1	0.1	0.1	0.1	0.1	0.1	0.0	0.1	0.1
Al ₂ O ₃	31.8	31.7	31.6	31.9	31.6	31.3	30.9	30.6	30.9	30.6	32.5	31.5	32.6	32.0	30.9	30.8
SiO ₂	50.6	50.0	50.2	49.6	50.9	50.0	50.0	50.8	50.0	50.8	50.6	49.9	49.4	51.0	51.2	51.1
K ₂ O	0.0	0.1	0.1	0.0	0.0	0.0	0.1	0.0	0.1	0.0	0.0	0.1	0.0	0.0	0.0	0.0
CaO	14.3	14.3	14.3	14.7	14.0	14.1	13.7	13.2	13.7	13.2	14.8	14.3	15.2	14.3	13.3	13.2
TiO ₂	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.1	0.0
Cr ₂ O ₃	0.3	0.2	0.2	0.2	0.2	0.2	0.1	0.2	0.1	0.2	0.3	0.2	0.3	0.3	0.3	0.4
MnO	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	-0.1	0.0
FeO	0.2	0.2	0.2	0.2	0.2	0.2	0.2	0.2	0.2	0.2	0.2	0.2	0.2	0.3	0.3	0.2
Total	100.9	100.2	100.2	100.0	100.8	99.3	98.7	99.2	98.7	99.2	102.1	99.9	101.1	101.6	100.2	99.9
Number of ions on the basis of 32 O																
Na	1.4	1.4	1.4	1.3	1.5	1.4	1.5	1.6	1.5	1.6	1.4	1.4	1.3	1.4	1.6	1.6
Mg	0.0	0.1	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0
Al	6.8	6.8	6.8	6.8	6.7	6.7	6.7	6.6	6.7	6.6	6.9	6.8	6.9	6.8	6.6	6.6
Si	9.1	9.1	9.1	9.1	9.2	9.2	9.2	9.3	9.2	9.3	9.0	9.1	8.9	9.1	9.3	9.3
K	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0
Ca	2.8	2.8	2.8	2.8	2.7	2.8	2.7	2.6	2.7	2.6	2.8	2.8	2.9	2.8	2.6	2.6
Ti	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0
Cr	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.1
Mn	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0
Fe ²⁺	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0
Z	16.0	16.0	16.0	16.0	16.0	16.0	16.0	16.0	16.0	16.0	16.0	16.0	16.0	16.0	16.0	16.0
X	4.2	4.2	4.2	4.1	4.2	4.2	4.2	4.2	4.2	4.2	4.2	4.2	4.2	4.2	4.2	4.2
An	66	66	66	68	64	66	64	61	64	61	67	66	70	65	61	61
Or	0	0	0	0	0	0	0	0	0	0	0	0	0	0	0	0
Ab	34	33	34	32	36	34	35	39	35	39	32	34	30	34	39	39

$$Z = \Sigma(\text{Si, Al}), X = \Sigma(\text{Na, Ca, K})$$

Table 2 Plagioclase analyses: sample B1 continued.

	17	18	19	20	21	22
Na ₂ O	3.8	3.6	3.1	3.2	2.9	3.5
MgO	0.1	0.1	0.1	0.1	0.0	0.0
Al ₂ O ₃	31.6	31.5	32.0	32.5	32.1	31.3
SiO ₂	50.3	49.6	48.6	49.7	48.0	49.1
K ₂ O	0.0	0.0	0.1	0.0	0.0	0.0
CaO	14.1	14.1	14.9	15.0	15.3	14.4
TiO ₂	0.1	0.0	0.0	0.1	0.0	0.1
Cr ₂ O ₃	0.4	0.5	0.1	0.1	0.3	0.1
MnO	0.0	0.0	-0.1	0.0	0.0	0.0
FeO	0.3	0.3	0.3	0.2	0.3	0.3
Total	100.7	99.8	99.1	100.9	99.0	98.8
Number of ions on the basis of 32 O						
Na	1.5	1.4	1.2	1.3	1.2	1.4
Mg	0.0	0.0	0.0	0.0	0.0	0.0
Al	6.7	6.8	6.9	6.9	7.0	6.8
Si	9.1	9.1	9.0	9.0	8.9	9.1
K	0.0	0.0	0.0	0.0	0.0	0.0
Ca	2.7	2.8	3.0	2.9	3.0	2.8
Ti	0.0	0.0	0.0	0.0	0.0	0.0
Cr	0.1	0.1	0.0	0.0	0.0	0.0
Mn	0.0	0.0	0.0	0.0	0.0	0.0
Fe ²⁺	0.0	0.1	0.0	0.0	0.0	0.0
Z	16.0	16.0	16.0	16.0	16.0	16.0
X	4.3	4.2	4.2	4.2	4.2	4.3
An	64	66	70	69	72	67
Or	0	0	0	0	0	0
Ab	36	34	29	30	28	33

$$Z = \Sigma(Si, Al), X = \Sigma(Na, Ca, K)$$

1-10 cumulus Feldspar
 11-14 resorbed Feldspar associated with
 clinopyroxene
 15-22 Feldspar rim around cpx altered to
 talc+tremolite

Table 3 Plagioclase analyses: sample B2.

	1	2	3	4	5	6	7	8	9	10	11	12	13	14	15	16	17	18	19
Na ₂ O	3.0	3.3	3.3	4.2	4.7	3.8	1.8	1.4	1.8	2.6	3.2	2.8	3.0	2.2	2.8	4.5	4.4	4.6	4.7
MgO	0.0	0.1	0.1	0.1	0.2	0.1	0.0	0.0	0.0	0.0	0.1	0.1	0.1	0.0	0.1	0.1	0.1	0.1	0.1
Al ₂ O ₃	32.2	31.3	31.3	30.5	29.3	31.1	19.1	19.0	18.7	19.3	32.7	33.1	32.5	33.3	33.1	30.2	30.9	30.6	30.1
SiO ₂	49.0	49.6	49.6	52.2	52.5	50.8	63.5	63.7	62.7	64.4	49.9	48.8	48.8	47.1	48.4	52.2	52.6	53.4	53.6
K ₂ O	0.2	0.3	0.2	0.3	0.2	0.2	14.1	15.0	14.0	13.3	0.1	0.1	0.1	0.0	0.1	0.1	0.1	0.2	0.3
CaO	15.2	14.3	14.3	13.0	11.9	13.7	0.1	0.0	0.1	0.2	15.3	16.0	15.5	16.6	16.0	12.8	13.1	12.8	12.3
TiO ₂	0.0	0.1	0.0	0.0	0.1	0.0	0.7	0.6	0.8	0.7	0.0	0.1	0.0	0.0	0.0	0.0	0.1	0.1	0.1
Cr ₂ O ₃	0.1	0.1	0.4	0.2	0.2	0.1	0.0	0.2	0.0	0.0	0.2	0.1	0.1	0.2	0.2	0.2	0.2	0.2	0.2
MnO	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0
FeO	0.2	0.2	0.3	0.2	0.2	0.2	0.1	0.0	0.0	0.1	0.1	0.2	0.1	0.2	0.1	0.2	0.2	0.2	0.2
Total	99.8	99.3	99.6	100.7	99.2	100.1	99.4	100.1	98.1	100.7	101.7	101.2	100.2	99.8	100.8	100.2	101.6	102.0	101.4
Number of ions on the basis of 32 O																			
Na	1.2	1.3	1.3	1.7	1.9	1.5	0.7	0.6	0.7	1.0	1.3	1.1	1.2	0.9	1.1	1.8	1.7	1.8	1.8
Mg	0.0	0.0	0.0	0.0	0.1	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0
Al	6.9	6.8	6.8	6.5	6.3	6.7	4.2	4.1	4.1	4.1	6.9	7.1	7.0	7.2	7.1	6.4	6.5	6.4	6.3
Si	9.0	9.1	9.1	9.4	9.6	9.2	11.7	11.8	11.8	11.7	9.0	8.8	8.9	8.7	8.8	9.4	9.4	9.5	9.6
K	0.0	0.1	0.1	0.1	0.0	0.0	3.3	3.5	3.3	3.1	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.1
Ca	3.0	2.8	2.8	2.5	2.3	2.7	0.0	0.0	0.0	0.0	2.9	3.1	3.0	3.3	3.1	2.5	2.5	2.4	2.3
Ti	0.0	0.0	0.0	0.0	0.0	0.0	0.1	0.1	0.1	0.1	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0
Cr	0.0	0.0	0.1	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0
Mn	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0
Fe ²⁺	0.1	0.1	0.1	0.0	0.1	0.1	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.1	0.0	0.0	0.1	0.0	0.1
Z	16.0	16.0	16.0	16.0	16.0	16.0	16.0	16.0	16.0	16.0	16.0	16.0	16.0	16.0	16.0	16.0	16.0	16.0	16.0
X	4.2	4.2	4.2	4.2	4.2	4.2	4.1	4.1	4.1	4.1	4.2	4.2	4.2	4.2	4.2	4.3	4.2	4.2	4.2
An	70	67	67	59	55	63	1	0	1	1	70	73	71	78	74	58	59	57	55
Or	1	1	1	2	1	1	82	86	82	74	0	0	0	0	0	1	1	1	1
Ab	29	32	32	39	44	36	18	14	18	25	30	26	28	21	26	41	40	42	43

 $Z = \Sigma(\text{Si}, \text{Al}), X = \Sigma(\text{Na}, \text{Ca}, \text{K})$

Table 4 Plagioclase analyses: sample B3.

	1	2	3	4	5	6	7	8
Na ₂ O	4.1	3.4	2.3	4.4	3.5	3.5	2.8	3.8
MgO	0.1	0.0	0.0	0.0	0.1	0.1	0.1	0.0
Al ₂ O ₃	30.7	32.3	33.9	31.0	32.5	32.5	33.5	32.2
SiO ₂	51.0	50.5	47.6	52.8	50.1	50.2	49.0	51.2
K ₂ O	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.1
CaO	13.3	14.6	16.9	13.3	14.9	14.9	15.7	14.2
TiO ₂	0.0	0.1	0.0	0.0	0.1	0.0	0.0	0.1
Cr ₂ O ₃	0.1	0.2	0.1	0.2	0.2	0.2	0.1	0.1
MnO	0.0	0.0	0.0	0.0	0.0	0.0	0.0	-0.1
FeO	0.1	0.1	0.1	0.1	0.2	0.2	0.2	0.1
Total	99.5	101.3	100.9	101.8	101.5	101.6	101.4	101.7
Number of ions on the basis of 32 O								
Na	1.6	1.4	0.9	1.7	1.4	1.4	1.4	1.5
Mg	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0
Al	6.6	6.8	7.3	6.5	6.9	6.9	6.8	6.8
Si	9.3	9.1	8.7	9.4	9.0	9.0	9.1	9.2
K	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0
Ca	2.6	2.8	3.3	2.5	2.9	2.9	2.8	2.7
Ti	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0
Cr	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0
Mn	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0
Fe ²⁺	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0
Z	16.0	16.0	16.0	16.0	16.0	16.0	16.0	16.0
X	4.2	4.2	4.2	4.3	4.2	4.2	4.2	4.2
An	61	67	78	59	68	68	67	65
Or	0	0	0	0	0	0	0	0
Ab	38	32	21	40	32	32	33	35

$$Z = \Sigma(Si, Al), X = \Sigma(Na, Ca, K)$$

Table 5 Plagioclase analyses: sample A4.

	1	2	3	4	5	6	7	8	9	10	11	12	13	14	15	16	17	18
Na ₂ O	3.6	3.5	3.4	3.5	3.6	3.5	3.6	3.5	3.6	3.6	3.5	3.6	3.6	3.5	1.6	1.6	1.6	1.5
MgO	0.1	0.1	0.0	0.1	0.0	0.0	0.0	0.1	0.1	0.1	0.0	0.2	0.0	0.0	0.1	0.0	0.1	0.0
Al ₂ O ₃	31.8	31.8	31.5	31.4	31.5	31.3	32.1	32.3	32.6	32.7	32.0	32.4	32.3	32.2	34.1	33.8	35.2	35.2
SiO ₂	50.0	50.3	49.4	49.5	49.8	49.5	50.5	49.9	50.6	50.8	50.1	50.3	50.8	50.6	45.2	45.1	46.2	46.0
K ₂ O	0.1	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.1	0.1	0.0	0.0	0.0	0.1	0.0	0.1	0.1
CaO	14.2	14.2	14.2	14.2	14.0	14.2	14.4	14.4	14.6	14.6	14.4	14.5	14.5	14.6	17.4	17.4	18.1	18.1
TiO ₂	0.1	0.0	0.0	0.0	0.1	0.0	0.0	0.0	0.0	0.1	0.0	0.0	0.1	-0.1	0.1	0.1	0.0	0.0
Cr ₂ O ₃	0.2	0.1	0.2	0.2	0.2	0.2	0.3	0.4	0.3	0.4	0.2	0.2	0.2	0.2	0.2	0.1	0.2	0.2
MnO	0.0	0.0	0.0	0.0	0.0	0.1	0.0	0.0	0.1	-0.1	0.0	0.0	0.0	0.0	0.0	0.1	0.0	0.0
FeO	0.1	0.2	0.1	0.1	0.2	0.1	0.2	0.3	0.1	0.2	0.2	0.2	0.1	0.1	0.1	0.1	0.2	0.1
Total	100.1	100.2	98.8	99.1	99.5	98.9	101.1	100.9	101.9	102.5	100.6	101.4	101.6	101.2	98.9	98.4	101.6	101.1
Number of ions on the basis of 32 O																		
Na	1.4	1.4	1.4	1.4	1.4	1.4	1.4	1.4	1.4	1.4	1.4	1.4	1.4	1.4	0.7	0.7	0.6	0.6
Mg	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0
Al	6.8	6.8	6.8	6.8	6.8	6.8	6.8	6.9	6.9	6.9	6.8	6.9	6.8	6.8	7.5	7.5	7.5	7.6
Si	9.1	9.1	9.1	9.1	9.1	9.1	9.1	9.0	9.0	9.0	9.1	9.0	9.1	9.1	8.4	8.4	8.4	8.4
K	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0
Ca	2.8	2.8	2.8	2.8	2.7	2.8	2.8	2.8	2.8	2.8	2.8	2.8	2.8	2.8	3.5	3.5	3.5	3.5
Ti	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0
Cr	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.1	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0
Mn	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0
Fe ²⁺	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0
Z	16.0	16.0	16.0	16.0	16.0	16.0	16.0	16.0	16.0	16.0	16.0	16.0	16.0	16.0	16.0	16.0	16.0	16.0
X	4.2	4.1	4.2	4.2	4.2	4.2	4.2	4.2	4.2	4.2	4.2	4.2	4.2	4.2	4.2	4.2	4.2	4.1
An	65	67	67	67	66	66	66	67	67	66	67	67	67	67	84	84	85	85
Or	0	0	0	0	0	0	0	0	0	0	0	0	0	0	0	0	0	0
Ab	34	34	33	34	34	34	34	33	33	34	33	33	33	33	16	16	15	15

$$Z = \Sigma(\text{Si, Al}), X = \Sigma(\text{Na, Ca, K})$$

Table 6 Plagioclase analyses: *sample A5*.

	1	2	3	4	5	6	7	8	9	10	11	14
Na ₂ O	3.7	3.6	3.6	3.8	4.2	3.8	3.9	3.7	3.9	4.2	11.1	11.4
MgO	0.1	0.1	0.2	0.1	0.1	0.1	0.1	0.1	0.1	0.1	0.2	0.2
Al ₂ O ₃	31.7	31.5	31.7	31.7	31.4	32.0	31.2	31.6	31.1	30.8	21.1	21.2
SiO ₂	50.5	50.5	50.4	51.0	52.3	51.9	51.3	50.8	51.2	51.3	67.5	66.9
K ₂ O	0.3	0.2	0.2	0.2	0.3	0.2	0.2	0.2	0.2	0.2	-0.1	0.0
CaO	14.2	14.1	14.1	14.1	13.6	14.1	13.6	13.9	13.5	13.0	1.2	1.3
TiO ₂	0.0	0.0	0.1	0.1	0.0	0.0	0.1	0.0	0.0	0.0	0.0	0.0
Cr ₂ O ₃	0.2	0.2	0.2	0.2	0.2	0.2	0.2	0.2	0.1	0.1	0.1	0.3
MnO	0.0	-0.1	0.0	0.0	0.0	0.0	0.0	-0.1	0.1	-0.1	0.0	0.1
FeO	0.3	0.2	0.2	0.2	0.2	0.3	0.3	0.2	0.2	0.2	0.2	0.3
Total	100.9	100.4	100.6	101.4	102.3	102.6	100.6	100.6	100.4	99.9	101.3	101.7
Number of ions on the basis of 32 O												
Na	1.4	1.4	1.4	1.5	1.6	1.5	1.5	1.4	1.6	1.7	4.2	4.3
Mg	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.1
Al	6.8	6.7	6.8	6.7	6.6	6.7	6.6	6.7	6.6	6.6	4.3	4.3
Si	9.1	9.2	9.1	9.2	9.3	9.2	9.3	9.2	9.3	9.3	11.6	11.5
K	0.1	0.0	0.1	0.1	0.1	0.1	0.0	0.0	0.0	0.0	0.0	0.0
Ca	2.7	2.7	2.7	2.7	2.6	2.7	2.6	2.7	2.6	2.5	0.2	0.2
Ti	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0
Cr	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0
Mn	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0
Fe ²⁺	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0
Z	16.0	16.0	16.0	16.0	16.0	16.0	16.0	16.0	16.0	16.0	16.0	15.9
X	4.2	4.2	4.2	4.2	4.3	4.2	4.2	4.2	4.2	4.2	4.4	4.5
An	65	65	65	64	60	64	63	64	62	60	5	5
Or	1	1	1	1	1	1	1	1	1	1	0	0
Ab	34	34	34	35	38	35	37	35	37	39	95	95

$$Z = \Sigma(\text{Si, Al}), X = \Sigma(\text{Na, Ca, K})$$

Table 7 Plagioclase analyses: sample C1.

	1	2	3	4	5	6	7	8	9	10	11	12	13	14	15	16	17	18	19	20
Na ₂ O	3.4	3.5	3.1	3.1	3.1	3.2	2.3	4.3	3.5	3.6	3.3	3.3	3.1	3.2	3.3	3.5	3.6	3.3	3.3	3.1
MgO	0.0	0.2	0.0	0.0	0.0	0.1	0.1	0.0	0.1	0.1	0.0	0.1	0.1	0.0	0.1	0.1	0.1	0.0	0.1	0.1
Al ₂ O ₃	31.7	31.6	31.2	31.3	31.3	31.4	31.5	28.7	31.6	31.4	32.1	32.5	32.6	32.3	31.1	31.6	31.4	32.1	32.5	32.6
SiO ₂	50.7	50.6	48.8	48.9	49.3	49.3	45.6	48.9	50.5	50.6	50.4	50.5	50.3	50.1	50.8	50.5	50.6	50.4	50.5	50.3
K ₂ O	0.3	0.4	0.3	0.3	0.3	0.3	0.0	0.0	0.3	0.3	0.2	0.3	0.2	0.2	0.7	0.3	0.3	0.2	0.3	0.2
CaO	14.5	14.5	14.5	14.4	14.6	14.5	20.6	16.3	14.1	14.2	14.9	14.9	15.2	15.1	15.3	14.1	14.2	14.9	14.9	15.2
TiO ₂	0.0	0.0	0.0	0.0	0.1	0.1	0.0	0.0	0.1	0.0	0.1	0.0	0.1	0.0	0.1	0.1	0.0	0.1	0.0	0.1
Cr ₂ O ₃	0.1	0.1	0.1	0.2	0.1	0.1	0.4	0.5	0.4	0.3	0.3	0.2	0.2	0.2	0.4	0.4	0.3	0.3	0.2	0.2
MnO	0.0	0.1	0.1	0.0	0.0	0.0	0.0	0.0	-0.1	0.0	0.0	0.0	-0.1	0.0	-0.1	-0.1	0.0	0.0	0.0	-0.1
FeO	0.2	0.2	0.2	0.2	0.1	0.2	0.3	0.2	0.3	0.2	0.3	0.3	0.2	0.3	0.3	0.3	0.2	0.3	0.3	0.2
Total	101.0	101.1	98.2	98.4	99.0	99.2	101.0	98.9	100.6	100.7	101.4	102.0	101.9	101.3	101.9	100.6	100.7	101.4	102.0	101.9
Number of ions on the basis of 32 O																				
Na	1.4	1.4	1.2	1.3	1.3	1.3	0.9	1.8	1.4	1.4	1.3	1.3	1.2	1.2	1.3	1.4	1.4	1.3	1.3	1.2
Mg	0.0	0.1	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0
Al	6.7	6.7	6.8	6.8	6.8	6.8	6.9	6.3	6.7	6.7	6.8	6.9	6.9	6.9	6.6	6.7	6.7	6.8	6.9	6.9
Si	9.1	9.1	9.1	9.1	9.1	9.1	8.5	9.1	9.2	9.2	9.1	9.0	9.0	9.0	9.1	9.2	9.2	9.1	9.0	9.0
K	0.1	0.1	0.1	0.1	0.1	0.1	0.0	0.0	0.1	0.1	0.0	0.1	0.1	0.1	0.2	0.1	0.1	0.0	0.1	0.1
Ca	2.8	2.8	2.9	2.9	2.9	2.9	4.1	3.3	2.7	2.7	2.9	2.9	2.9	2.9	2.9	2.7	2.7	2.9	2.9	2.9
Ti	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0
Cr	0.0	0.0	0.0	0.0	0.0	0.0	0.1	0.1	0.1	0.0	0.0	0.0	0.0	0.0	0.1	0.1	0.0	0.0	0.0	0.0
Mn	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0
Fe ²⁺	0.0	0.0	0.0	0.0	0.0	0.0	0.1	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0
Z	16.0	16.0	16.0	16.0	15.9	16.0	15.5	15.5	16.0	16.0	16.0	16.0	16.0	16.0	15.9	16.0	16.0	16.0	16.0	16.0
X	4.2	4.3	4.2	4.2	4.2	4.2	5.0	5.0	4.2	4.2	4.2	4.2	4.2	4.2	4.4	4.2	4.2	4.2	4.2	4.2
An	66	66	69	68	69	68	81	65	65	65	68	68	70	69	67	65	65	68	68	70
Or	2	2	2	2	2	2	0	0	1	1	1	1	1	1	4	1	1	1	1	1
Ab	32	32	30	30	30	31	19	35	33	34	31	31	29	30	29	33	34	31	31	29

 $Z = \Sigma(\text{Si, Al}), X = \Sigma(\text{Na, Ca, K})$

Table 7 continued. Plagioclase analyses: sample C1

	21	22	23	24	25	26
Na ₂ O	3.2	3.3	10.1	10.2	10.4	10.7
MgO	0.0	0.1	0.1	0.0	0.1	0.1
Al ₂ O ₃	32.3	31.1	22.7	22.6	21.6	21.5
SiO ₂	50.1	50.8	65.5	65.3	66.0	66.1
K ₂ O	0.2	0.7	0.0	0.0	0.0	0.0
CaO	15.1	15.3	3.1	3.0	1.9	1.7
TiO ₂	0.0	0.1	0.0	0.0	0.1	0.0
Cr ₂ O ₃	0.2	0.4	0.1	0.1	0.4	0.4
MnO	0.0	-0.1	0.0	0.0	0.0	0.0
FeO	0.3	0.3	0.2	0.3	0.2	0.2
Total	101.3	101.9	101.7	101.4	100.6	100.8
Number of ions on the basis of 32 O						
Na	1.2	1.3	3.8	3.8	3.9	4.0
Mg	0.0	0.0	0.0	0.0	0.0	0.0
Al	6.9	6.6	4.6	4.6	4.4	4.4
Si	9.0	9.1	11.3	11.3	11.4	11.5
K	0.1	0.2	0.0	0.0	0.0	0.0
Ca	2.9	2.9	0.6	0.6	0.4	0.3
Ti	0.0	0.0	0.0	0.0	0.0	0.0
Cr	0.0	0.1	0.0	0.0	0.1	0.1
Mn	0.0	0.0	0.0	0.0	0.0	0.0
Fe ²⁺	0.0	0.0	0.0	0.0	0.0	0.0
Z	16.0	15.9	15.9	15.9	16.0	0.0
X	4.2	4.4	4.4	4.4	4.3	16.0
An	69	67	13	13	8	7
Or	1	4	0	0	0	0
Ab	30	29	87	87	92	93

$$Z = \Sigma(\text{Si, Al}), X = \Sigma(\text{Na, Ca, K})$$

Table 8 Plagioclase analyses: sample C5.

	1	2	3	4	5	6	7	8	9	10	11	12	13	14	15	16	17	18
Na ₂ O	3.5	2.9	2.9	3.1	3.1	3.2	3.1	2.2	2.4	1.5	2.3	2.3	3.4	3.6	3.3	3.4	3.4	2.9
MgO	0.1	0.1	0.1	0.1	0.1	0.1	0.1	0.1	0.1	0.1	0.1	0.1	0.0	0.1	0.0	0.1	0.1	0.1
Al ₂ O ₃	31.5	31.6	31.6	31.6	31.8	31.7	31.2	33.1	32.8	34.6	33.1	33.1	31.7	31.5	31.9	31.9	31.9	32.1
SiO ₂	50.5	48.3	48.3	49.0	49.3	49.9	48.6	47.3	47.9	46.1	47.9	47.9	50.4	50.6	49.9	50.6	50.8	49.0
K ₂ O	0.3	0.3	0.3	0.3	0.3	0.3	0.3	0.2	0.3	0.4	0.2	0.2	0.3	0.3	0.3	0.3	0.4	0.3
CaO	14.1	14.9	14.9	14.5	14.6	14.4	14.6	16.2	15.8	17.6	16.1	16.1	14.4	14.1	14.6	14.2	14.2	15.0
TiO ₂	0.0	0.0	0.0	0.0	0.0	0.1	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.1	0.0	0.0	0.0	0.0
Cr ₂ O ₃	0.2	0.1	0.1	0.2	0.3	0.1	0.2	0.3	0.4	0.6	0.4	0.4	0.3	0.3	0.3	0.4	0.3	0.3
MnO	0.0	0.0	0.0	0.0	0.0	-0.1	-0.1	0.1	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0
FeO	0.2	0.2	0.2	0.2	0.2	0.2	0.1	0.3	0.3	0.4	0.4	0.4	0.2	0.3	0.3	0.3	0.4	0.3
Total	100.4	98.2	98.2	99.0	99.7	100.1	98.1	99.7	100.1	101.3	100.5	100.5	100.6	100.9	100.7	101.2	101.6	100.0
Number of ions on the basis of 32 O																		
Na	1.4	1.2	1.2	1.2	1.2	1.3	1.3	0.9	1.0	0.6	0.9	0.9	1.3	1.4	1.3	1.4	1.3	1.2
Mg	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0
Al	6.7	6.9	6.9	6.9	6.9	6.8	6.8	7.2	7.1	7.4	7.1	7.1	6.8	6.7	6.8	6.8	6.8	6.9
Si	9.2	9.0	9.0	9.0	9.0	9.1	9.0	8.7	8.8	8.4	8.8	8.8	9.1	9.1	9.1	9.1	9.1	9.0
K	0.1	0.1	0.1	0.1	0.1	0.1	0.1	0.0	0.1	0.1	0.1	0.1	0.1	0.1	0.1	0.1	0.1	0.1
Ca	2.7	3.0	3.0	2.9	2.9	2.8	2.9	3.2	3.1	3.4	3.1	3.1	2.8	2.7	2.8	2.7	2.7	2.9
Ti	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0
Cr	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.1	0.1	0.1	0.1	0.0	0.0	0.0	0.1	0.0	0.0
Mn	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0
Fe ²⁺	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.1	0.1	0.1	0.1	0.0	0.0	0.1	0.0	0.1	0.0
Z	16.0	16.0	16.0	16.0	16.0	16.0	16.0	16.0	16.0	16.0	16.0	16.0	16.0	16.0	16.0	16.0	16.0	16.0
X	4.2	4.2	4.2	4.2	4.2	4.2	4.2	4.1	4.1	4.1	4.1	4.1	4.2	4.2	4.2	4.2	4.2	4.2
An	65	71	71	69	69	67	69	77	75	83	76	76	66	65	68	66	66	70
Or	2	1	1	2	2	2	2	1	1	2	1	1	2	2	2	2	2	2
Ab	33	28	28	30	30	31	30	22	24	15	23	23	32	34	31	32	32	28

 $Z = \Sigma(\text{Si, Al}), X = \Sigma(\text{Na, Ca, K})$

Table 9 Pyroxene analyses: Sample A1.

	1	2	3	4	5	6	7	8	9	10	11	12	13	14	15	16	17	18	19	20	21	22
Na ₂ O	0.5	0.5	0.4	0.5	0.5	0.5	0.5	0.5	0.5	0.5	0.5	0.5	0.5	0.5	0.5	0.5	0.5	0.5	0.5	0.5	0.5	0.5
MgO	33.7	33.2	33.4	32.8	33.3	32.6	32.4	32.6	32.7	32.7	33.1	32.6	32.8	32.9	33.2	33.2	32.6	32.8	32.7	33.3	33.0	33.4
Al ₂ O ₃	1.1	1.0	1.5	0.8	0.9	1.1	1.2	1.1	1.2	1.3	1.0	1.0	0.8	1.1	1.0	1.0	1.2	1.2	1.0	0.9	1.0	0.9
SiO ₂	57.0	56.7	56.9	56.2	56.5	56.2	56.0	56.0	56.1	56.0	56.3	55.9	56.2	56.4	56.8	56.8	56.3	56.4	55.8	56.9	56.3	56.9
K ₂ O	-0.1	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	-0.1	0.0	0.0	0.0	0.0	0.0	0.0	0.0	-0.1	-0.1	0.0	-0.1	-0.1
CaO	0.7	0.8	0.6	0.7	0.6	0.8	0.8	0.7	0.7	0.7	0.7	0.8	0.7	0.8	0.7	0.7	0.7	0.8	0.8	0.5	0.8	0.5
TiO ₂	0.1	0.1	0.1	0.1	0.1	0.2	0.1	0.1	0.2	0.2	0.1	0.1	0.1	0.1	0.1	0.1	0.1	0.1	0.2	0.2	0.2	0.2
Cr ₂ O ₃	0.4	0.4	0.5	0.4	0.5	0.5	0.4	0.4	0.4	0.5	0.5	0.4	0.4	0.5	0.5	0.5	0.4	0.5	0.8	0.4	0.5	0.4
MnO	0.2	0.2	0.2	0.2	0.2	0.2	0.2	0.2	0.2	0.2	0.2	0.2	0.2	0.2	0.3	0.3	0.2	0.2	0.2	0.3	0.2	0.2
FeO	8.2	8.6	8.4	8.8	8.3	8.8	9.0	8.7	8.7	8.8	8.2	8.5	8.8	8.4	8.6	8.6	8.4	8.9	8.4	8.7	8.5	8.6
NiO	0.1	0.0	0.1	0.0	0.1	0.1	0.1	0.1	0.1	0.1	0.1	0.0	0.0	0.1	0.1	0.1	0.1	0.0	0.0	0.1	0.0	0.1
Total	102.0	101.5	102.1	100.4	101.0	101.0	100.7	100.4	100.8	100.9	100.6	100.0	100.4	100.9	101.7	101.7	100.4	101.3	100.3	101.7	100.8	101.6
Number of ions on the basis of 6 O																						
Na	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0
Mg	1.7	1.7	1.7	1.7	1.7	1.7	1.7	1.7	1.7	1.7	1.7	1.7	1.7	1.7	1.7	1.7	1.7	1.7	1.7	1.7	1.7	1.7
Al	0.0	0.0	0.1	0.0	0.0	0.0	0.0	0.0	0.0	0.1	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0
Si	2.0	2.0	1.9	2.0	2.0	2.0	2.0	2.0	2.0	1.9	2.0	2.0	2.0	2.0	2.0	2.0	2.0	2.0	1.9	2.0	2.0	2.0
K	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0
Ca	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0
Ti	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0
Cr	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0
Mn	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0
Fe ²⁺	0.2	0.2	0.2	0.3	0.2	0.3	0.3	0.3	0.3	0.3	0.2	0.3	0.3	0.2	0.2	0.2	0.2	0.3	0.2	0.2	0.2	0.2
Ni	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0
Z	2.0	2.0	2.0	2.0	2.0	2.0	2.0	2.0	2.0	2.0	2.0	2.0	2.0	2.0	2.0	2.0	2.0	2.0	2.0	2.0	2.0	2.0
X	2.0	2.0	2.0	2.0	2.0	2.0	2.0	2.0	2.0	2.0	2.0	2.0	2.0	2.0	2.0	2.0	2.0	2.0	2.0	2.0	2.0	2.0
Wo	1	1	1	1	1	2	2	1	1	1	1	1	1	2	1	1	1	1	1	1	1	1
En	87	86	87	86	87	85	85	86	86	86	87	86	86	86	86	86	86	86	86	86	86	87
Fs	12	13	12	13	12	13	13	13	13	13	12	13	13	12	13	13	12	13	12	13	12	12

$$Z = \Sigma(Al, Si), X = \Sigma(Na, Mg, K, Ca, Ti, Cr, Mn, Fe^{2+}, Ni)$$

Table 10 Pyroxene analyses: Sample B1.

	1	2	3	4	5	6	7	8	9	10	11	12	13	14	15	16	17	18	19	20	21	22
Na ₂ O	0.5	0.5	0.6	0.5	0.5	0.5	0.6	0.5	0.5	0.5	0.4	0.6	0.6	0.6	0.6	0.5	0.5	0.5	0.5	0.6	0.5	0.6
MgO	33.1	33.0	33.1	31.8	32.2	31.3	32.4	31.8	33.2	33.2	33.0	16.8	16.6	16.7	17.2	17.1	17.1	16.9	16.8	16.7	17.0	16.7
Al ₂ O ₃	1.1	0.8	1.0	1.3	1.1	1.2	1.1	1.3	1.0	1.0	1.3	1.8	2.1	2.0	1.7	1.4	1.6	1.7	1.5	1.6	1.8	1.8
SiO ₂	56.7	55.9	56.1	55.2	55.7	54.8	56.2	55.2	56.2	56.2	56.4	52.6	52.4	52.2	53.2	53.1	53.1	52.5	52.8	52.2	53.3	52.5
K ₂ O	0.0	0.0	0.0	-0.1	0.0	-0.1	0.0	-0.1	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0
CaO	0.7	0.4	0.6	0.6	0.8	0.8	0.7	0.6	0.7	0.6	0.7	23.3	22.8	22.9	23.2	23.5	23.2	24.1	23.4	23.0	23.4	23.1
TiO ₂	0.2	0.1	0.1	0.1	0.1	0.1	0.1	0.1	0.1	0.2	0.2	0.4	0.3	0.3	0.3	0.3	0.2	0.4	0.3	0.3	0.3	0.3
Cr ₂ O ₃	0.6	0.4	0.6	0.4	0.4	0.4	0.4	0.4	0.6	0.5	0.5	1.1	1.2	1.2	1.1	1.0	1.1	1.1	1.1	1.1	1.1	1.2
MnO	0.2	0.1	0.3	0.3	0.2	0.2	0.2	0.3	0.3	0.2	0.2	0.1	0.1	0.1	0.1	0.1	0.1	0.0	0.0	0.1	0.1	0.1
FeO	8.7	8.0	8.2	9.6	9.5	9.6	9.3	9.6	8.3	8.0	8.5	3.3	3.6	3.4	3.1	3.1	3.3	2.9	3.2	3.1	3.2	3.2
NiO	0.1	0.0	0.1	0.1	0.1	0.0	0.0	0.1	0.0	0.1	0.1	0.0	0.1	0.1	0.0	0.1	0.1	0.1	0.0	0.1	0.1	0.1
Total	101.9	99.1	100.5	99.8	100.6	98.8	101.1	99.8	100.9	100.4	101.2	99.9	99.6	99.4	100.5	100.2	100.3	100.2	99.6	98.8	100.7	99.5
Number of ions on the basis of 6 O																						
Na	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0
Mg	1.7	1.7	1.7	1.7	1.7	1.7	1.7	1.7	1.7	1.7	1.7	0.9	0.9	0.9	0.9	0.9	0.9	0.9	0.9	0.9	0.9	0.9
Al	0.0	0.0	0.0	0.1	0.0	0.0	0.0	0.1	0.0	0.0	0.1	0.1	0.1	0.1	0.1	0.1	0.1	0.1	0.1	0.1	0.1	0.1
Si	1.9	2.0	2.0	1.9	1.9	2.0	2.0	1.9	1.9	2.0	1.9	1.9	1.9	1.9	1.9	1.9	1.9	1.9	1.9	1.9	1.9	1.9
K	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0
Ca	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.9	0.9	0.9	0.9	0.9	0.9	0.9	0.9	0.9	0.9	0.9
Ti	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0
Cr	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0
Mn	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0
Fe ²⁺	0.3	0.2	0.2	0.3	0.3	0.3	0.3	0.3	0.2	0.2	0.2	0.1	0.1	0.1	0.1	0.1	0.1	0.1	0.1	0.1	0.1	0.1
Ni	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0
Z	2.0	2.0	2.0	2.0	2.0	2.0	2.0	2.0	2.0	2.0	2.0	2.0	2.0	2.0	2.0	2.0	2.0	2.0	2.0	2.0	2.0	2.0
X	2.0	2.0	2.0	2.0	2.0	2.0	2.0	2.0	2.0	2.0	2.0	2.0	2.0	2.0	2.0	2.0	2.0	2.0	2.0	2.0	2.0	2.0
Wo	1	1	1	1	1	2	1	1	1	1	1	47	47	47	47	47	47	48	47	47	47	47
En	86	87	87	85	85	84	85	85	87	87	86	47	47	48	48	48	48	47	47	48	48	48
Fs	13	12	12	14	14	14	14	14	12	12	12	5	6	5	5	5	5	5	5	5	5	5

 $Z = \Sigma(Al, Si)$, $X = \Sigma(Na, Mg, K, Ca, Ti, Cr, Mn, Fe^{2+}, Ni)$

Table 10 continued. Pyroxene analyses: Sample B1

	23	24	25	26	27
Na ₂ O	0.6	0.6	0.6	0.5	0.5
MgO	16.5	17.3	16.9	16.8	16.7
Al ₂ O ₃	1.8	1.6	1.8	1.5	1.5
SiO ₂	51.8	53.7	52.9	52.7	52.2
K ₂ O	0.0	0.0	0.0	0.0	0.0
CaO	22.5	23.5	23.2	23.5	22.9
TiO ₂	0.3	0.3	0.3	0.3	0.3
Cr ₂ O ₃	1.1	1.0	1.2	0.9	0.8
MnO	0.0	0.1	0.0	0.1	0.1
FeO	3.6	3.1	3.3	2.8	3.2
NiO	0.1	0.1	0.1	0.0	0.1
Total	98.2	101.1	100.1	99.0	98.3
Number of ions on the basis of 6 O					
Na	0.0	0.0	0.0	0.0	0.0
Mg	0.9	0.9	0.9	0.9	0.9
Al	0.1	0.1	0.1	0.1	0.1
Si	1.9	1.9	1.9	1.9	1.9
K	0.0	0.0	0.0	0.0	0.0
Ca	0.9	0.9	0.9	0.9	0.9
Ti	0.0	0.0	0.0	0.0	0.0
Cr	0.0	0.0	0.0	0.0	0.0
Mn	0.0	0.0	0.0	0.0	0.0
Fe ²⁺	0.1	0.1	0.1	0.1	0.1
Ni	0.0	0.0	0.0	0.0	0.0
Z	2.0	2.0	2.0	2.0	2.0
X	2.0	2.0	2.0	2.0	2.0
Wo	47	47	47	48	47
En	48	48	48	48	48
Fs	6	5	5	5	5

$Z = \Sigma(Al, Si)$, $X = \Sigma(Na, Mg, K, Ca, Ti, Cr, Mn, Fe^{2+}, Ni)$

Table 11 Pyroxene analyses: Sample B2 (1-4) and B3 (5-9).

	1	2	3	4	5	6	7	8	9
Na ₂ O	0.5	0.5	0.5	0.4	0.5	0.5	0.6	0.4	0.5
MgO	32.5	33.0	33.1	33.3	17.3	16.6	16.9	16.8	33.1
Al ₂ O ₃	1.6	1.7	1.7	1.7	1.6	1.7	1.5	1.5	1.5
SiO ₂	54.8	56.0	56.2	56.4	53.6	52.3	52.9	52.7	56.8
K ₂ O	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	-0.1
CaO	0.2	0.4	0.3	0.4	23.9	23.1	23.7	23.7	0.4
TiO ₂	0.2	0.2	0.2	0.2	0.3	0.4	0.3	0.3	0.2
Cr ₂ O ₃	0.5	0.5	0.6	0.6	1.0	1.1	1.0	0.9	0.5
MnO	0.2	0.2	0.2	0.2	0.1	0.1	0.0	0.1	0.2
FeO	8.0	8.3	8.2	8.3	3.2	3.4	3.1	3.0	8.5
NiO	0.0	0.1	0.1	0.1	0.1	0.1	0.1	0.1	0.0
Total	98.5	100.8	101.1	101.5	101.5	99.2	100.0	99.4	101.8
Number of ions on the basis of 6 O									
Na	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0
Mg	1.7	1.7	1.7	1.7	0.9	0.9	0.9	0.9	1.7
Al	0.1	0.1	0.1	0.1	0.1	0.1	0.1	0.1	0.1
Si	1.9	1.9	1.9	1.9	1.9	1.9	1.9	1.9	2.0
K	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0
Ca	0.0	0.0	0.0	0.0	0.9	0.9	0.9	0.9	0.0
Ti	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0
Cr	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0
Mn	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0
Fe ²⁺	0.2	0.2	0.2	0.2	0.1	0.1	0.1	0.1	0.2
Ni	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0
Z	2.0	2.0	2.0	2.0	2.0	2.0	2.0	2.0	2.0
X	2.0	2.0	2.0	2.0	2.0	2.0	2.0	2.0	2.0
Wo	0	1	1	1	47	47	48	48	1
En	88	87	87	87	48	47	47	47	87
Fs	12	12	12	12	5	5	5	5	13

$$Z = \Sigma(Al, Si), X = \Sigma(Na, Mg, K, Ca, Ti, Cr, Mn, Fe^{2+}, Ni)$$

Table 12 Pyroxene analyses: Sample A4.

	1	2	3	4	5	6	7	8	9	10	11	12	13	14	15	16	17	18	19	20	21
Na ₂ O	0.6	0.6	0.5	0.6	0.7	0.6	0.5	0.5	0.5	0.6	0.6	0.6	0.5	0.7	0.6	0.7	0.5	0.4	0.4	0.6	0.5
MgO	31.6	31.8	31.6	16.7	17.0	16.5	16.7	16.8	16.7	17.2	17.2	16.9	16.8	16.8	16.8	17.0	17.0	17.0	16.8	17.1	16.6
Al ₂ O ₃	1.0	1.0	1.3	2.0	1.9	1.7	1.8	1.8	1.7	1.8	1.8	1.6	1.8	1.7	1.8	1.7	1.6	1.4	1.5	1.5	1.7
SiO ₂	56.4	56.2	56.4	52.6	52.6	52.9	52.9	53.4	53.3	54.0	54.1	53.4	53.3	53.1	53.1	53.4	53.6	53.8	53.6	53.7	52.8
K ₂ O	0.0	0.0	-0.1	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	-0.1	0.0	0.1	0.0	0.0	0.0
CaO	0.7	0.7	1.0	22.1	21.6	22.6	22.6	22.8	22.7	22.9	22.8	22.9	23.0	22.4	22.6	23.2	23.3	23.1	23.2	23.4	22.8
TiO ₂	0.1	0.1	0.1	0.3	0.3	0.3	0.3	0.3	0.2	0.3	0.3	0.3	0.3	0.3	0.2	0.3	0.3	0.2	0.2	0.2	0.3
Cr ₂ O ₃	0.5	0.4	0.5	1.0	1.0	1.1	1.1	1.2	1.1	1.2	1.2	1.1	1.0	0.9	1.0	0.9	0.9	0.8	1.1	1.1	1.0
MnO	0.2	0.3	0.2	0.1	0.1	0.2	0.1	0.1	0.1	0.1	0.1	0.0	0.1	0.1	0.1	0.1	0.2	0.1	0.1	0.0	0.1
FeO	10.6	10.6	10.4	4.6	4.9	4.4	4.4	4.4	4.3	4.6	4.7	4.4	4.1	4.5	4.4	4.1	3.7	4.6	4.0	3.7	4.4
NiO	0.0	0.0	0.1	0.1	0.1	0.0	0.1	0.1	0.0	0.0	0.0	0.2	0.1	0.0	0.1	0.0	0.1	0.1	0.1	0.0	0.1
Total	101.7	101.7	102.0	100.0	100.2	100.2	100.6	101.2	100.7	102.5	102.7	101.4	101.0	100.4	100.6	101.3	101.1	101.4	101.1	101.3	100.1
Number of ions on the basis of 6 O																					
Na	0.0	0.0	0.0	0.0	0.1	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.1	0.0	0.1	0.0	0.0	0.0	0.0	0.0
Mg	1.6	1.6	1.6	0.9	0.9	0.9	0.9	0.9	0.9	0.9	0.9	0.9	0.9	0.9	0.9	0.9	0.9	0.9	0.9	0.9	0.9
Al	0.0	0.0	0.1	0.1	0.1	0.1	0.1	0.1	0.1	0.1	0.1	0.1	0.1	0.1	0.1	0.1	0.1	0.1	0.1	0.1	0.1
Si	2.0	2.0	2.0	1.9	1.9	1.9	1.9	1.9	1.9	1.9	1.9	1.9	1.9	1.9	1.9	1.9	1.9	1.9	1.9	1.9	1.9
K	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0
Ca	0.0	0.0	0.0	0.9	0.8	0.9	0.9	0.9	0.9	0.9	0.9	0.9	0.9	0.9	0.9	0.9	0.9	0.9	0.9	0.9	0.9
Ti	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0
Cr	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0
Mn	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0
Fe ²⁺	0.3	0.3	0.3	0.1	0.2	0.1	0.1	0.1	0.1	0.1	0.1	0.1	0.1	0.1	0.1	0.1	0.1	0.1	0.1	0.1	0.1
Ni	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0
Z	2.0	2.0	2.0	2.0	2.0	2.0	2.0	2.0	2.0	2.0	2.0	2.0	2.0	2.0	2.0	2.0	2.0	2.0	2.0	2.0	2.0
X	2.0	2.1	2.0	2.0	2.0	2.0	2.0	2.0	2.0	2.0	2.0	2.0	2.0	2.0	2.0	2.0	2.0	2.0	2.0	2.0	2.0
Wo	1	1	2	45	44	46	46	46	46	45	45	46	46	45	46	46	47	46	47	47	46
En	83	83	83	47	48	47	47	47	47	47	48	47	47	47	47	47	47	47	47	47	47
Fs	16	16	15	7	8	7	7	7	7	7	7	7	7	7	7	6	6	7	6	6	7

$$Z = \Sigma(Al, Si), X = \Sigma(Na, Mg, K, Ca, Ti, Cr, Mn, Fe^{2+}, Ni)$$

Table 12 continued Pyroxene analyses: *Sample A4*

	22	23	24	25	26
Na ₂ O	0.4	0.4	0.4	0.5	0.5
MgO	16.6	16.5	16.6	16.6	16.5
Al ₂ O ₃	1.9	1.6	1.7	1.7	1.6
SiO ₂	52.9	52.3	52.7	52.2	52.0
K ₂ O	0.0	0.0	0.0	0.0	0.0
CaO	22.7	23.0	22.9	22.5	22.5
TiO ₂	0.2	0.3	0.2	0.2	0.2
Cr ₂ O ₃	0.9	1.1	1.0	1.0	0.9
MnO	0.1	0.1	0.2	0.0	0.1
FeO	4.5	4.0	4.1	4.2	4.4
NiO	0.1	0.1	0.2	0.1	0.1
Total	100.3	99.4	99.8	99.0	98.7
Number of ions on the basis of 6 O					
Na	0.0	0.0	0.0	0.0	0.0
Mg	0.9	0.9	0.9	0.9	0.9
Al	0.1	0.1	0.1	0.1	0.1
Si	1.9	1.9	1.9	1.9	1.9
K	0.0	0.0	0.0	0.0	0.0
Ca	0.9	0.9	0.9	0.9	0.9
Ti	0.0	0.0	0.0	0.0	0.0
Cr	0.0	0.0	0.0	0.0	0.0
Mn	0.0	0.0	0.0	0.0	0.0
Fe ²⁺	0.1	0.1	0.1	0.1	0.1
Ni	0.0	0.0	0.0	0.0	0.0
Z	2.0	2.0	2.0	2.0	2.0
X	2.0	2.0	2.0	2.0	2.0
Wo	46	47	47	46	46
En	47	47	47	47	47
Fs	7	6	6	7	7

 $Z = \Sigma(Al, Si)$, $X = \Sigma(Na, Mg, K, Ca, Ti, Cr, Mn, Fe^{2+}, Ni)$
Table 13 Pyroxene analyses: *Sample C1(1-7) and C5 (8-9)*.

	1	2	3	4	5	6	7	8	9
Na ₂ O	0.5	0.5	0.5	0.5	0.5	0.5	0.4	0.5	0.4
MgO	33.9	33.7	33.5	33.7	33.8	34.2	34.2	17.1	17.0
Al ₂ O ₃	1.3	1.4	1.4	1.4	1.2	1.2	1.3	1.7	1.9
SiO ₂	56.5	56.4	56.6	56.8	56.6	56.8	56.8	53.1	52.5
K ₂ O	0.0	-0.1	-0.1	0.0	0.0	-0.1	0.0	0.0	0.0
CaO	0.4	0.5	0.5	0.5	0.5	0.4	0.4	24.1	24.0
TiO ₂	0.2	0.2	0.1	0.2	0.2	0.2	0.2	0.3	0.4
Cr ₂ O ₃	0.9	0.8	0.7	0.9	0.9	0.7	0.6	1.2	1.3
MnO	0.2	0.3	0.3	0.2	0.2	0.2	0.2	0.1	0.0
FeO	7.8	8.1	8.3	7.9	7.8	7.6	7.7	2.9	3.0
NiO	0.1	0.1	0.1	0.1	0.1	0.1	0.1	0.0	0.1
Total	101.6	101.8	101.9	101.9	101.8	101.8	101.8	101.0	100.4
Number of ions on the basis of 6 O									
Na	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0
Mg	1.7	1.7	1.7	1.7	1.7	1.7	1.7	0.9	0.9
Al	0.1	0.1	0.1	0.1	0.0	0.0	0.1	0.1	0.1
Si	1.9	1.9	1.9	1.9	1.9	1.9	1.9	1.9	1.9
K	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0
Ca	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.9	0.9
Ti	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0
Cr	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0
Mn	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0
Fe ²⁺	0.2	0.2	0.2	0.2	0.2	0.2	0.2	0.1	0.1
Ni	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0
Z	2.0	2.0	2.0	2.0	2.0	2.0	2.0	2.0	2.0
X	2.0	2.0	2.0	2.0	2.0	2.0	2.0	2.0	2.0
Wo	1	1	1	1	1	1	1	48	48
En	88	87	87	88	88	88	88	47	47
Fs	11	12	12	11	11	11	11	5	5

 $Z = \Sigma(Al, Si)$, $X = \Sigma(Na, Mg, K, Ca, Ti, Cr, Mn, Fe^{2+}, Ni)$

Table 14 Selected phlogopite analyses.

	1	2	3	4	5	6	7
Na ₂ O	0.4	1.0	1.7	1.7	0.3	0.3	0.2
MgO	22.3	20.6	24.3	23.8	21.9	22.0	21.8
Al ₂ O ₃	15.6	15.8	17.9	16.9	14.3	14.2	14.3
SiO ₂	38.3	38.0	39.8	40.2	38.5	38.6	38.7
K ₂ O	10.4	9.3	7.9	7.9	9.8	9.7	9.8
CaO	0.0	0.0	0.0	0.0	0.0	0.1	0.1
TiO ₂	4.3	1.6	0.6	1.4	4.5	4.6	5.2
Cr ₂ O ₃	1.8	1.5	1.2	1.4	2.0	2.0	1.9
MnO	0.0	0.0	0.0	0.1	0.0	0.0	0.1
FeO	3.0	7.4	2.6	2.8	4.1	4.0	4.0
NiO	0.2	0.3	0.2	0.2	0.2	0.2	0.2
Cl	0.1	0.4	0.1	0.1	0.5	0.5	0.5
Total	96.5	95.5	96.3	96.2	95.7	95.8	96.2
Number of ions on the basis of 24 O							
Na	0.1	0.3	0.5	0.5	0.1	0.1	0.1
Mg	4.8	4.5	5.1	5.0	4.7	4.7	4.7
Al	2.6	2.7	3.0	2.8	2.4	2.4	2.5
Si	5.5	5.5	5.6	5.7	5.6	5.6	5.6
K	1.9	1.7	1.4	1.4	1.8	1.8	1.8
Ca	0.0	0.0	0.0	0.0	0.0	0.0	0.0
Ti	0.5	0.2	0.1	0.1	0.5	0.5	0.6
Cr	0.2	0.2	0.1	0.2	0.2	0.2	0.2
Mn	0.0	0.0	0.0	0.0	0.0	0.0	0.0
Fe ²⁺	0.4	0.9	0.3	0.3	0.5	0.5	0.5
Ni	0.0	0.0	0.0	0.0	0.0	0.0	0.0
OH	3.3	4.0	3.4	3.5	3.7	3.6	3.2
Cl	0.0	0.1	0.0	0.0	0.1	0.1	0.1
Σ(Al,Si)	8.0	8.0	8.0	8.0	8.0	8.0	8.0
Σ(Ca,Na,K)	2.0	2.0	2.0	1.9	1.9	1.9	1.9
Σ(Fe,Mg,Al,Ti,Cr)	6.0	6.0	6.2	6.1	6.0	6.0	6.1
Σ(OH,Cl)	3.3	4.1	3.4	3.6	3.8	3.7	3.3

Table 15 Selected amphibole analyses: Edenitic hornblende (1,3,5-9) & tremolite (2,4)

	1	2	3	4	5	6	7	8	9
Na ₂ O	2.6	0.6	2.5	0.6	2.6	2.7	2.7	2.8	2.4
MgO	17.8	16.5	17.9	15.9	17.3	17.7	17.7	17.5	18.5
Al ₂ O ₃	11.4	4.5	10.1	3.1	11.0	11.1	11.3	11.6	11.7
SiO ₂	44.4	52.2	45.2	53.0	44.5	45.1	44.2	44.3	46.0
K ₂ O	0.6	0.1	0.4	0.0	0.2	0.2	0.3	0.3	0.2
CaO	12.4	11.2	12.3	12.0	12.2	12.5	12.3	12.3	12.5
TiO ₂	2.4	0.1	2.3	0.1	2.3	2.4	2.7	2.6	0.9
Cr ₂ O ₃	2.2	0.7	2.0	0.4	2.1	2.1	2.0	2.0	1.7
MnO	0.0	0.4	0.0	0.5	0.1	0.0	0.0	0.1	0.0
FeO	4.4	11.0	4.2	11.0	5.0	4.6	4.4	4.6	4.0
NiO	0.1	0.0	0.1	0.1	0.2	0.1	0.1	0.1	0.1
Total	98.0	97.3	97.0	96.8	97.4	98.5	97.6	98.1	97.9
Number of cations based on 24 O									
Na	0.8	0.2	0.8	0.2	0.8	0.8	0.8	0.9	0.7
Mg	3.8	3.5	3.8	3.4	3.7	3.8	3.8	3.7	3.9
Al	1.9	0.8	1.7	0.5	1.9	1.9	1.9	2.0	2.0
Si	6.4	7.4	6.4	7.5	6.4	6.5	6.3	6.4	6.5
K	0.1	0.0	0.1	0.0	0.0	0.0	0.0	0.0	0.0
Ca	1.9	1.7	1.9	1.8	1.9	1.9	1.9	1.9	1.9
Ti	0.3	0.0	0.2	0.0	0.3	0.3	0.3	0.3	0.1
Cr	0.2	0.1	0.2	0.0	0.2	0.2	0.2	0.2	0.2
Mn	0.0	0.1	0.0	0.1	0.0	0.0	0.0	0.0	0.0
Fe ²⁺	0.5	1.3	0.5	1.3	0.6	0.5	0.5	0.6	0.5
Ni	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0
OH	1.9	2.5	2.9	3.0	2.5	1.5	2.3	1.8	2.0
Σ(Al,Si)	8.0	8.0	8.0	8.0	8.0	8.0	8.0	8.0	8.0
Σ(Ca,Na,K)	2.8	1.9	2.7	2.0	2.7	2.8	2.8	2.8	2.7
Σ(Fe,Mg,Al,Ti,Cr)	5.2	5.1	4.9	4.8	5.0	5.3	5.0	5.2	5.2

Table 16 Selected pumpellyite analyses.

	1	2	3	4	5	6
Na ₂ O	0.3	0.2	0.1	0.1	0.1	0.3
MgO	1.0	4.3	3.7	3.3	3.6	3.9
Al ₂ O ₃	27.2	24.5	26.1	26.6	26.3	25.8
SiO ₂	37.4	37.4	37.6	37.7	37.7	37.6
K ₂ O	-0.0	0.0	0.0	-0.0	0.0	-0.0
CaO	23.1	23.0	23.5	23.5	23.8	23.5
TiO ₂	0.0	0.1	0.0	0.0	0.1	0.0
Cr ₂ O ₃	0.2	1.4	0.4	0.3	0.3	0.4
MnO	0.4	0.2	0.4	0.6	0.2	0.1
FeO	4.3	1.9	1.9	1.9	2.1	1.8
NiO	-0.0	0.1	-0.0	0.0	0.0	0.1
Total	93.9	93.1	93.7	94.1	94.2	93.4
Number of ions based on 14 O						
Na	0.1	0.0	0.0	0.0	0.0	0.0
Mg	0.1	0.5	0.4	0.4	0.4	0.5
Al	2.6	2.3	2.5	2.5	2.5	2.4
Si	3.0	3.0	3.0	3.0	3.0	3.0
K	0.0	0.0	0.0	0.0	0.0	0.0
Ca	2.0	2.0	2.0	2.0	2.1	2.0
Ti	0.0	0.0	0.0	0.0	0.0	0.0
Cr	0.0	0.1	0.0	0.0	0.0	0.0
Mn	0.0	0.0	0.0	0.0	0.0	0.0
Fe ²⁺	0.3	0.1	0.1	0.1	0.1	0.1
Ni	0.0	0.0	0.0	0.0	0.0	0.0
OH	3.3	3.7	3.4	3.2	3.1	3.5
Si	3.0	3.0	3.0	3.0	3.0	3.0
Σ(Ca,Na,K)	2.0	2.0	2.0	2.0	2.1	2.1
Σ(Fe ²⁺ Mg,Al)	3.0	3.1	3.1	3.1	3.1	3.0

Table 17 Selected epidote (1-3) and prehnite (4) analyses.

	1	2	3	4
Na	0.2	0.4	0.1	0.2
Mg	0.3	4.6	0.8	0.0
Al	29.7	26.8	29.6	24.3
Si	38.5	40.1	38.3	43.9
K	0.0	0.0	0.0	0.0
Ca	23.3	23.9	23.4	27.2
Ti	0.1	0.1	0.0	0.0
Cr	0.5	0.3	0.6	0.0
Mn	0.2	0.1	0.3	0.1
Fe ²⁺	4.8	2.2	4.5	0.2
Ni	0.1	0.0	0.0	0.1
Total	97.5	98.2	97.4	96.2
Number of ions based on 12.5 O for epidote and 24 O for prehnite				
Na	0.0	0.1	0.0	0.1
Mg	0.0	0.5	0.1	0.0
Al	2.6	2.3	2.6	4.0
Si	2.9	3.0	2.8	6.1
K	0.0	0.0	0.0	0.0
Ca	1.9	1.9	1.9	4.0
Ti	0.0	0.0	0.0	0.0
Cr	0.0	0.0	0.0	0.0
Mn	0.0	0.0	0.0	0.0
Fe ²⁺	0.3	0.1	0.3	0.0
Ni	0.0	0.0	0.0	0.0
OH	1.2	0.9	1.3	3.7
Σ(Si,Al)	3.0	3.0	3.0	6.0
Σ(Ca,Na,K)	1.9	2.0	1.9	4.1
Σ(Fe ²⁺ Mg,Al)	2.8	3.0	2.8	4.0

Table 18 Selected chlorite (1-2), septechlorite (3-5), serpentine (6-8) and talc (9-10) analyses.

	1	2	3	4	5	6	7	8	9	10
Na	0.7	0.7	0.5	0.6	0.6	0.4	0.5	0.3	0.5	0.5
Mg	21.6	21.3	23.7	23.4	23.3	38.6	36.0	41.0	29.6	28.8
Al	12.9	12.2	0.0	0.0	0.1	0.2	0.0	-0.1	6.6	4.7
Si	36.5	37.8	55.2	54.4	55.3	42.6	43.8	43.5	51.1	54.1
K	0.0	0.0	0.0	0.0	0.1	0.0	-0.1	0.0	0.1	0.1
Ca	4.7	5.1	0.6	0.6	0.5	0.2	0.3	0.0	0.1	0.4
Ti	0.1	0.0	0.0	0.0	0.0	-0.1	0.0	0.0	0.0	0.0
Cr	2.3	2.2	0.1	0.0	0.0	0.0	0.0	0.0	0.0	0.0
Mn	0.3	0.4	0.2	0.1	0.1	0.1	0.1	0.0	0.1	0.1
Fe	11.7	11.8	11.1	11.0	11.8	4.6	6.0	1.8	5.8	5.1
Ni	0.1	0.1	0.5	0.5	0.4	0.2	0.1	0.4	0.0	0.0
Total	90.8	91.6	91.8	90.7	92.3	86.9	86.8	86.8	94.8	94.6
Number of ions based on 36 O for chlorite, 18 O for septechlorite, 9 O for serpentine, and 24 O for talc										
Na	0.3	0.3	0.1	0.1	0.1	0.0	0.1	0.0	0.2	0.1
Mg	6.5	6.5	3.5	3.4	3.4	2.7	2.5	2.8	5.8	5.6
Al	3.1	2.9	0.0	0.0	0.0	0.0	0.0	0.0	1.0	0.7
Si	7.4	7.7	5.4	5.3	5.4	2.0	2.1	2.0	6.7	7.0
K	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0
Ca	1.0	1.1	0.1	0.1	0.1	0.0	0.0	0.0	0.0	0.1
Ti	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0
Cr	0.4	0.4	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0
Mn	0.0	0.1	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0
Fe ²⁺	2.0	2.0	0.9	0.9	1.0	0.2	0.2	0.1	0.6	0.6
Ni	0.0	0.0	0.1	0.1	0.1	0.0	0.0	0.0	0.0	0.0
OH	12.4	11.4	5.3	6.0	5.1	4.1	4.1	4.1	5.2	5.3
Σ(Al,Si)	8.0	8.0	4.0	4.0	4.0	2.0	2.0	2.0	7.7	7.7
Σ(Fe,Mg,Al,Na,Ca,K)	12.4	11.4	6.6	6.4	6.0	3.0	2.8	2.9	6.4	6.1

APPENDIX F

Electron-microprobe analyses of selected base-metal sulphide grains. Analytical conditions and minimum detection limits are discussed in Chapter 4.

- Table 1** Chalcopyrite compositions from sample A1.
- Table 2** Chalcopyrite compositions from sample B4.
- Table 3** Chalcopyrite compositions from sample C1.
- Table 4** Pyrrhotite compositions from sample A1.
- Table 5** Pyrrhotite compositions from sample B4
- Table 6** Pyrite compositions from sample A1.
- Table 7** Pyrite compositions from sample B4.
- Table 8** Pyrite compositions from sample C1.
- Table 9** Pyrite compositions from sample C2.
- Table 10** Millerite compositions from sample C1.
- Table 11** Millerite compositions from sample C2.
- Table 12** Pentlandite compositions from sample A1.
- Table 13** Pentlandite compositions from sample B4.
- Table 14** Pentlandite compositions from sample C1.
- Table 15** Siegenite compositions from samples C1 and C2.

APPENDIX G

Electron-microprobe analyses of selected PGE mineral grains.

Table 1 Selected electron-microprobe analyses of cooperite.

Table 2 Selected electron-microprobe analyses of braggite and vysotskite.

Table 3 Selected electron-microprobe analyses of malanite.

Table 4 Selected electron-microprobe analyses of Pt-Fe alloy

Table 5 Electron-microprobe analyses of atheneite (1), kotulskite (2,4), and moncheite (3).

Table 3 Selected electron-microprobe analyses of malanite.

	1	2	3	4	5	6	7	8	9	10
As	0.1	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0
S	26.7	26.0	27.4	26.9	27.0	26.4	26.5	26.4	26.5	26.7
Ru	0.0	0.0	0.0	0.0	0.0	0.0	0.1	0.0	0.0	0.0
Rh	13.3	6.9	7.1	8.6	14.0	9.4	4.6	6.5	10.7	9.0
Pd	0.0	0.0	0.1	0.0	0.1	0.1	0.1	0.1	0.0	0.1
Pt	41.6	47.1	38.6	46.1	40.6	42.0	42.8	47.8	42.2	43.2
Fe	1.0	0.8	3.9	1.6	0.8	0.9	2.1	0.9	1.6	1.5
Co	2.6	3.8	3.0	2.7	3.1	3.3	6.3	3.1	3.7	3.5
Ni	0.5	1.8	7.0	2.8	0.8	6.4	3.4	2.3	0.5	2.8
Cu	12.4	12.6	11.3	12.8	12.9	11.8	12.4	12.5	12.9	12.4
Total	98.3	99.2	98.4	101.7	99.5	100.3	98.3	99.8	98.6	99.3
Atomic proportions										
As	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0
S	4.0	4.0	4.0	4.0	4.0	4.0	4.0	4.0	4.0	4.0
Ru	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0
Rh	0.6	0.3	0.3	0.4	0.7	0.4	0.2	0.3	0.5	0.4
Pd	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0
Pt	1.0	1.2	0.9	1.1	1.0	1.1	1.1	1.2	1.1	1.1
Fe	0.1	0.1	0.3	0.1	0.1	0.1	0.2	0.1	0.1	0.1
Co	0.2	0.3	0.2	0.2	0.3	0.3	0.5	0.3	0.3	0.3
Ni	0.0	0.2	0.6	0.2	0.1	0.5	0.3	0.2	0.0	0.2
Cu	1.0	1.1	0.9	1.0	1.1	1.0	1.0	1.0	1.1	1.0

Table 4 Selected electron-microprobe analyses of Pt-Fe alloy.

	1	2	3
S	0.1	0.1	0.3
Ru	0.0	0.0	0.0
Rh	0.2	0.3	0.5
Pd	0.1	0.2	0.2
Pt	89.0	86.7	86.4
Pb	0.4	0.4	0.5
Bi	0.0	0.2	0.1
Fe	9.4	9.3	9.7
Co	0.0	0.0	0.0
Ni	0.1	0.1	0.6
Cu	0.2	0.8	0.3
Total	99.5	98.2	98.7
Atomic proportions			
S	0.0	0.0	0.0
Ru	0.0	0.0	0.0
Rh	0.0	0.0	0.0
Pd	0.0	0.0	0.0
Pt	2.6	2.9	2.8
Pb	0.0	0.0	0.0
Bi	0.0	0.0	0.0
Fe	1.2	1.1	1.0
Co	0.0	0.0	0.0
Ni	0.0	0.0	0.0
Cu	0.1	0.0	0.1
ΣPGE	2.7	2.9	2.8
Σ(Fe,Cu, Ni,Co)	1.3	1.1	1.1
Ni	0.0	0.0	0.0
Cu	0.0	0.0	0.0
ΣPGE	2.8	0.9	1.0
Σ(Te,Bi,Hg,As)	1.2	2.0	3.0

* high total + low total

Table 5 Electron-microprobe analyses of atheneite (1), kotulskite (2,4), and moncheite (3).

	1*	2	3+	4
As	22.0	0.0	0.1	0.0
S	0.0	0.0	0.1	0.1
Ru	0.0	0.0	0.0	0.0
Rh	0.0	0.0	0.6	0.0
Pd	64.0	38.9	8.4	35.7
Te	0.1	43.8	53.4	47.4
Pt	0.1	0.0	26.2	5.8
Hg	15.9	0.0	0.0	0.0
Pb	0.0	0.0	0.1	0.2
Bi	0.0	15.1	7.6	10.0
Fe	0.4	0.5	0.5	1.0
Ni	0.0	0.0	0.1	0.0
Cu	0.0	0.0	0.1	0.1
Total	102.4	98.5	97.2	100.2
Atomic proportions				
As	1.2	0.0	0.0	0.0
S	0.0	0.0	0.0	0.0
Ru	0.0	0.0	0.0	0.0
Rh	0.0	0.0	0.0	0.0
Pd	2.4	0.9	0.3	0.8
Te	0.0	0.9	1.8	0.9
Pt	0.0	0.0	0.6	0.1
Hg	0.3	0.0	0.0	0.0
Pb	0.0	0.0	0.0	0.0
Bi	0.0	0.2	0.2	0.1
Fe	0.0	0.0	0.0	0.0
Ni	0.0	0.0	0.0	0.0
Cu	0.0	0.0	0.0	0.0
ΣPGE	2.8	0.9	1.0	0.9
Σ(Te,Bi,Hg,As)	1.2	2.0	3.0	2.0

* high total + low total

APPENDIX H

Mineralogical data for milled feed and flotation products

- Table 1** Chromite grain-size distributions in fourteen UG2 chromitite samples milled to 80% <75µm (reported in area per cent).
- Table 2** Silicate grain-size distributions in fourteen UG2 chromitite samples milled to 80% <75µm (reported in area per cent).
- Table 3** Mode of occurrence of PGE minerals in fourteen samples of UG2 chromitite milled to 80% <75 µm.
- Table 4** Chromite grain-size distribution in flotation products of sample C1.
- Table 5** Silicate grain-size distribution in flotation products of sample C1.
- Table 6** Relative proportions of silicate minerals in flotation products of selected samples determined by image analysis. Reported in volume %.
- Table 7** Liberation characteristics of base-metal sulphide grains in flotation products of selected samples (reported in area %).
- Table 8** Chalcopyrite recovery to three flotation concentrates, assuming a 100% total recovery.
- Table 9** Pentlandite recovery to three flotation concentrates, assuming a 100% total recovery.
- Table 10** Pyrrhotite recovery to three flotation concentrates, assuming a 100% total recovery.
- Table 11** Pyrite recovery to three flotation concentrates, assuming a 100% total recovery.
- Table 12** Millerite recovery to three flotation concentrates, assuming a 100% total recovery.
- Table 13** Relative proportions of **liberated** grains of different PGE minerals in flotation concentrates of samples A1, A3, A4, B4, A5, C1 and C2.

Table 1 Chromite grain-size distributions in fourteen UG2 chromitite samples milled to 80% <math><75\mu\text{m}</math> (reported in area per cent).

Size class (μm)	A2	B2	A1	B1	A3	B3	A4	B4	A5	C1	C2	C3	C5
0.0-17.0	12	16	15	14	16	14	15	13	21	14	14	13	16
17.0-36.0	29	32	31	31	34	33	34	31	33	29	33	32	31
36.0-55.0	21	18	19	21	20	20	22	20	17	20	22	21	20
55.0-74.1	17	16	14	15	16	14	15	15	12	15	14	15	15
74.1-93.1	12	11	11	11	9	10	9	12	9	11	11	11	10
93.1-112.1	7	5	6	6	3	6	5	6	5	6	5	6	5
112.1-131.1	1	2	2	2	1	2	1	2	2	3	2	2	2
131.1-150.1	0	0	1	0	0	1	0	0	1	1	0	0	0
150.1-169.1	0	0	0	0	0	0	0	0	0	0	0	0	0
169.1-188.1	0	0	0	0	0	0	0	0	0	0	0	0	0
188.1-207.2	0	0	0	0	0	0	0	0	0	0	0	0	0
107.2-226.2	0	0	0	0	0	0	0	0	0	0	0	0	0
>226.2	0	0	0	0	0	0	0	0	0	0	0	0	0
Median	44	38	40	41	36	38	37	41	32	42	39	41	38

Table 2 Silicate grain-size distributions in fourteen UG2 chromitite samples milled to 80% <math><75\mu\text{m}</math> (reported in area per cent).

Size class (μm)	A2	B2	A1	B1	A3	B3	A4	B4	A5	C1	C2	C3	C5
0.0-17.0	23	28	28	27	29	30	32	29	36	30	30	31	27
17.0-36.0	23	23	24	24	24	25	24	22	22	20	25	22	26
36.0-55.0	22	20	21	19	22	19	21	19	16	20	21	20	23
55.0-74.1	19	16	16	15	15	15	13	16	13	15	15	16	14
74.1-93.1	9	10	8	9	7	8	7	9	8	9	7	9	7
93.1-112.1	3	3	2	4	2	3	2	3	4	5	2	2	2
112.1-131.1	0	1	0	1	0	1	1	1	1	2	0	1	1
131.1-150.1	0	0	0	0	0	0	0	0	1	0	0	0	0
150.1-169.1	0	0	0	0	0	0	0	0	0	0	0	0	0
169.1-188.1	0	0	0	0	0	0	0	0	0	0	0	0	0
188.1-207.2	0	0	0	0	0	0	0	0	0	0	0	0	0
107.2-226.2	0	0	0	0	0	0	0	0	0	0	0	0	0
>226.2	0	0	0	0	0	0	0	0	0	0	0	0	0
Median	39	35	34	35	34	32	31	35	30	36	32	34	34

Table 3 Mode of occurrence of PGE minerals in fourteen samples of UG2 chromitite milled to 80% <75 μ m.

	A2	B2	A1	B1	A3	B3	A4	B4	A5	C1	C2	C3	C4	C5
<i>number of PGEMs</i>	185	193	619	187	217	190	211	217	216	236	185	191	203	183
	% Number of grains													
<i>Liberated PGEM</i>	51	46	57	59	62	48	59	67	33	59	68	57	65	71
<i>PGEM locked in BMS</i>	15	16	14	11	8	14	11	9	4	8	1	3	5	3
<i>PGEM locked in oxide</i>	2	4	3	2	3	4	0	2	4	5	5	3	5	4
<i>PGEM locked in silicate</i>	3	4	1	2	5	6	4	4	20	12	4	18	10	5
<i>Grain boundary BMS/Gangue</i>	3	6	4	5	5	3	3	2	16	6	9	7	6	4
<i>Grain edge of BMS</i>	18	20	16	18	10	17	17	7	1	3	5	4	3	4
<i>Grain edge of gangue</i>	8	5	5	4	6	7	5	10	22	7	8	8	6	8
	Volume %													
<i>Liberated PGEM</i>	49	48	57	64	60	57	76	66	48	62	61	54	61	76
<i>PGEM locked in BMS</i>	15	14	13	6	6	10	5	11	1	4	0	2	3	1
<i>PGEM locked in oxide</i>	1	2	3	3	1	3	0	2	3	8	8	2	5	2
<i>PGEM locked in silicate</i>	4	1	1	1	15	5	2	1	22	9	1	17	6	3
<i>Grain boundary BMS/Gangue</i>	3	2	3	3	9	2	2	1	9	8	12	9	14	9
<i>Grain edge of BMS</i>	17	31	18	18	5	18	12	15	0	4	8	3	4	1
<i>Grain edge of gangue</i>	11	3	5	4	5	4	3	5	17	4	10	12	7	8

Table 4 Chromite grain-size distribution in flotation products of sample C1 (reported in area per cent).

Size class (μm)	RC1	RC2	RC3-5	RT
0.0-17.0	99.2	96.7	98.8	4.9
17.0-36.0	0.8	1.2	1.2	23.2
36.0-55.0	0.0	2.1	0.0	24
55.0-74.1	0	0	0	18.4
74.1-93.1	0	0	0	14.1
93.1-112.1	0	0	0	9.2
112.1-131.1	0	0	0	3.6
131.1-150.1	0	0	0	2.1
150.1-169.1	0	0	0	0.1
169.1-188.1	0	0	0	0
188.1-207.2	0	0	0	0
207.2-226.2	0	0	0	0
>225	0	0	0	0.4
	100	100	100	100

RC1= rougher concentrate 1, RC2=rougher concentrate 2, RC3-5=combined rougher concentrate 3 to 5, RT=rougher tailings

Table 5 Silicate grain-size distributions in flotation products of sample C1 (reported in area per cent).

Size class (μm)	RC1	RC2	RC3-5	RT
0.0-17.0	85	81.9	88	9
17.0-36.0	12.3	13.6	6.4	31.4
36.0-55.0	2.7	4.4	5.8	25.1
55.0-74.1	0	0	0	14.8
74.1-93.1	0	0	0	11.6
93.1-112.1	0	0	0	5.5
112.1-131.1	0	0	0	1.4
131.1-150.1	0	0	0	0.8
150.1-169.1	0	0	0	0.5
169.1-188.1	0	0	0	0
188.1-207.2	0	0	0	0
207.2-226.2	0	0	0	0
>225	0	0	0	0

RC1= rougher concentrate 1, RC2=rougher concentrate 2, RC3-5=combined rougher concentrate 3 to 5, RT=rougher tailings

Table 6 Relative proportions of silicate minerals in flotation products of selected samples determined by image analysis. Reported in volume %.

A1	RC1	RC2	RC3-5	RT
<i>Ca-Al-silicate*</i>	63	62	58	57
<i>Mg-Fe-silicate⁺</i>	31	32	35	40
<i>Phlogopite</i>	2	2	2	1
<i>Clinopyroxene</i>	1	1	1	1
<i>Quartz</i>	0	0	0	0
<i>Chlorite</i>	0	1	1	1
<i>Amphibole</i>	2	2	2	0
<i>Albite</i>	0	0	0	0
<i>K-Al-silicate^o</i>	0	0	0	0
A3	RC1	RC2	RC3-5	RT
<i>Ca-Al-silicate*</i>	62	63	58	69
<i>Mg-Fe-silicate⁺</i>	33	31	36	28
<i>Phlogopite</i>	1	2	1	1
<i>Clinopyroxene</i>	1	1	1	0
<i>Quartz</i>	0	0	0	0
<i>Chlorite</i>	1	1	1	0
<i>Amphibole</i>	0	1	2	0
<i>Albite</i>	1	1	1	0
<i>K-Al-silicate^o</i>	0	1	1	1
B3	RC1	RC2	RC3-5	RT
<i>Ca-Al-silicate*</i>	50	43	43	65
<i>Mg-Fe-silicate⁺</i>	42	48	51	31
<i>Phlogopite</i>	1	1	0	2
<i>Clinopyroxene</i>	3	3	4	1
<i>Quartz</i>	0	0	0	0
<i>Chlorite</i>	1	1	1	0
<i>Amphibole</i>	5	3	2	1
<i>Albite</i>	0	0	0	0
<i>K-Al-silicate^o</i>	0	0	0	0
A4	RC1	RC2	RC3-5	RT
<i>Ca-Al-silicate*</i>	21	22	21	24
<i>Mg-Fe-silicate⁺</i>	48	50	47	39
<i>Phlogopite</i>	0	0	0	0
<i>Clinopyroxene</i>	19	19	20	29
<i>Quartz</i>	1	1	1	0
<i>Chlorite</i>	2	2	2	0
<i>Amphibole</i>	6	6	9	1
<i>Albite</i>	0	0	0	0
<i>K-Al-silicate^o</i>	0	0	0	0

Table 6 continued Relative proportions of silicate minerals in flotation products of selected samples determined by image analysis. Reported in volume %.

B4	RC1	RC2	RC3-5	RT
<i>Ca-Al-silicate*</i>	33	38	41	68
<i>Mg-Fe-silicate⁺</i>	54	49	45	19
<i>Phlogopite</i>	0	0	1	1
<i>Clinopyroxene</i>	8	6	7	8
<i>Quartz</i>	0	0	0	0
<i>Chlorite</i>	1	3	2	1
<i>Amphibole</i>	3	4	4	1
<i>Albite</i>	0	0	0	0
<i>K-Al-silicate^o</i>	0	0	0	0
A5	RC1	RC2	RC3-5	RT
<i>Ca-Al-silicate*</i>	16	16	23	50
<i>Mg-Fe-silicate⁺</i>	68	53	44	16
<i>Phlogopite</i>	0	0	0	1
<i>Clinopyroxene</i>	8	8	11	7
<i>Quartz</i>	2	1	0	1
<i>Chlorite</i>	3	15	14	16
<i>Amphibole</i>	2	6	6	6
<i>Albite</i>	2	1	1	4
<i>K-Al-silicate^o</i>	0	0	0	0
C1	RC1	RC2	RC3-5	RT
<i>Ca-Al-silicate*</i>	8	10	18	66
<i>Mg-Fe-silicate⁺</i>	70	64	51	12
<i>Phlogopite</i>	0	0	0	0
<i>Clinopyroxene</i>	4	5	5	4
<i>Quartz</i>	2	2	2	4
<i>Chlorite</i>	12	14	17	6
<i>Amphibole</i>	3	3	3	2
<i>Albite</i>	1	1	2	4
<i>K-Al-silicate^o</i>	0	0	1	1
C4	RC1	RC2	RC3-5	RT
<i>Ca-Al-silicate*</i>	27	25	28	77
<i>Mg-Fe-silicate⁺</i>	63	66	64	18
<i>Phlogopite</i>	0	0	1	1
<i>Clinopyroxene</i>	1	1	1	0
<i>Quartz</i>	1	0	1	1
<i>Chlorite</i>	5	5	4	1
<i>Amphibole</i>	3	3	3	0
<i>Albite</i>	1	0	1	1
<i>K-Al-silicate^o</i>	0	0	0	0

RC1= rougher concentrate 1, RC2=rougher concentrate 2, RC3-5=combined rougher concentrate 3 to 5, RT=rougher tailings

* Predominantly plagioclase. Also includes pumpellyite, prehnite and epidote, especially in samples from area C.

⁺ Predominantly orthopyroxene, minor talc, and rarely serpentine and olivine

^o K-feldspar and sericite

Table 7 Liberation characteristics of base-metal sulphide grains in flotation products of selected samples (reported in area %).

RC1	<i>A1</i>	<i>A3</i>	<i>A4</i>	<i>B4</i>	<i>A5</i>	<i>C1</i>	<i>C2</i>
<i>0.0-0.2</i>	0	4	1	1	9	5	3
<i>0.2-0.4</i>	1	4	0	0	16	5	2
<i>0.4-0.6</i>	0	1	0	0	3	5	6
<i>0.6-0.8</i>	0	0	0	0	0	0	0
<i>0.8-1.0</i>	98	91	99	99	72	84	89
RC2	<i>A1</i>	<i>A3</i>	<i>A4</i>	<i>B4</i>	<i>A5</i>	<i>C1</i>	<i>C2</i>
<i>0.0-0.2</i>	2	4	0	1	12	14	5
<i>0.2-0.4</i>	2	0	0	1	7	0	5
<i>0.4-0.6</i>	0	1	0	0	0	0	0
<i>0.6-0.8</i>	0	0	0	0	0	0	0
<i>0.8-1.0</i>	96	94	99	98	80	86	89
RC3-5	<i>A1</i>	<i>A3</i>	<i>A4</i>	<i>B4</i>	<i>A5</i>	<i>C1</i>	<i>C2</i>
<i>0.0-0.2</i>	8	13	4	3	28	24	19
<i>0.2-0.4</i>	0	16	2	2	3	0	1
<i>0.4-0.6</i>	0	0	0	0	0	0	0
<i>0.6-0.8</i>	0	0	0	0	0	0	0
<i>0.8-1.0</i>	92	71	94	94	69	76	80

RC1= rougher concentrate 1, *RC2*=rougher concentrate 2, *RC3-5*=combined rougher concentrates 3 to 5

Table 8 Chalcopyrite recovery to three flotation concentrates, assuming a 100% total recovery.

A2	<i>cpy content g/t</i>	<i>cpy content g/t cum.</i>	<i>cpy distribution. %</i>	<i>cpy distribution % cum.</i>
RC-1	0.38	0.38	67.87	67.87
RC-2	0.08	0.22	16.85	84.71
RC-3-5	0.05	0.14	15.29	100.00
B2	<i>cpy content g/t</i>	<i>cpy content g/t cum.</i>	<i>cpy distribution. %</i>	<i>cpy distribution % cum.</i>
RC-1	0.32	0.32	69.54	69.54
RC-2	0.05	0.16	13.69	83.23
RC-3-5	0.03	0.10	16.77	100.00
A1	<i>cpy content g/t</i>	<i>cpy content g/t cum.</i>	<i>cpy distribution. %</i>	<i>cpy distribution % cum.</i>
RC-1	0.35	0.35	72.47	72.47
RC-2	0.10	0.25	13.66	86.13
RC-3-5	0.05	0.17	13.87	100.00
B1	<i>cpy content g/t</i>	<i>cpy content g/t cum.</i>	<i>cpy distribution. %</i>	<i>cpy distribution % cum.</i>
RC-1	0.65	0.65	82.57	82.57
RC-2	0.09	0.43	7.52	90.09
RC-3-5	0.06	0.26	9.91	100.00
A3	<i>cpy content g/t</i>	<i>cpy content g/t cum.</i>	<i>cpy distribution. %</i>	<i>cpy distribution % cum.</i>
RC-1	0.6463	0.65	65.56	65.56
RC-2	0.1699	0.41	16.69	82.25
RC-3-5	0.10	0.26	17.75	100.00
B3	<i>cpy content g/t</i>	<i>cpy content g/t cum.</i>	<i>cpy distribution. %</i>	<i>cpy distribution % cum.</i>
RC-1	0.82	0.82	72.34	72.34
RC-2	0.16	0.47	15.04	87.38
RC-3-5	0.06	0.25	12.62	100.00
A4	<i>cpy content g/t</i>	<i>cpy content g/t cum.</i>	<i>cpy distribution. %</i>	<i>cpy distribution % cum.</i>
RC-1	0.78	0.78	59.43	59.43
RC-2	0.24	0.50	20.74	80.17
RC-3-5	0.11	0.29	19.83	100.00
B4	<i>cpy content g/t</i>	<i>cpy content g/t cum.</i>	<i>cpy distribution. %</i>	<i>cpy distribution % cum.</i>
RC-1	1.34	1.34	69.30	69.30
RC-2	0.24	0.75	14.98	84.28
RC-3-5	0.09	0.35	15.72	100.00
A5	<i>cpy content g/t</i>	<i>cpy content g/t cum.</i>	<i>cpy distribution. %</i>	<i>cpy distribution % cum.</i>
RC-1	0.26	0.26	63.65	63.65
RC-2	0.06	0.16	16.04	79.69
RC-3-5	0.04	0.10	20.31	100.00
C1	<i>cpy content g/t</i>	<i>cpy content g/t cum.</i>	<i>cpy distribution. %</i>	<i>cpy distribution % cum.</i>
RC-1	0.28	0.28	54.79	54.79
RC-2	0.08	0.17	18.89	73.68
RC-3-5	0.05	0.10	26.32	100.00

Table 8 continued Chalcopyrite recovery to three flotation concentrates, assuming a 100% total recovery.

C2	<i>cpy content</i> g/t	<i>cpy content</i> g/t cum.	<i>cpy distribution.</i> %	<i>cpy distribution</i> % cum.
RC-1	0.22	0.22	53.39	53.39
RC-2	0.07	0.14	18.67	72.06
RC-3-5	0.04	0.08	27.94	100.00
C3	<i>cpy content</i> g/t	<i>cpy content</i> g/t cum.	<i>cpy distribution.</i> %	<i>cpy distribution</i> % cum.
RC-1	0.20	0.20	50.59	50.59
RC-2	0.07	0.13	19.94	70.53
RC-3-5	0.04	0.08	29.47	100.00
C4	<i>cpy content</i> g/t	<i>cpy content</i> g/t cum.	<i>cpy distribution.</i> %	<i>cpy distribution</i> % cum.
RC-1	0.12	0.12	48.37	48.37
RC-2	0.04	0.08	22.00	70.37
RC-3-5	0.03	0.05	29.63	100.00
C5	<i>cpy content</i> g/t	<i>cpy content</i> g/t cum.	<i>cpy distribution.</i> %	<i>cpy distribution</i> % cum.
RC-1	0.26	0.26	53.67	53.67
RC-2	0.07	0.15	19.15	72.82
RC-3-5	0.04	0.09	27.18	100.00
Comp.	<i>cpy content</i> g/t	<i>cpy content</i> g/t cum.	<i>cpy distribution.</i> %	<i>cpy distribution</i> % cum.
RC-1	0.47	0.47	67.56	67.56
RC-2	0.11	0.29	15.52	83.08
RC-3-5	0.06	0.17	16.92	100.00

Table 9 Pentlandite recovery to three flotation concentrates, assuming a 100% total recovery.

A2	<i>pn content</i> g/t	<i>pn content</i> g/t cum.	<i>pn distribution.</i> %	<i>pn distribution</i> % cum.
RC-1	1.11	1.11	63.60	63.60
RC-2	0.45	0.76	29.19	92.79
RC-3-5	0.07	0.44	7.21	100.00
B2	<i>pn content</i> g/t	<i>pn content</i> g/t cum.	<i>pn distribution.</i> %	<i>pn distribution</i> % cum.
RC-1	0.63	0.63	51.33	51.33
RC-2	0.28	0.43	32.56	83.89
RC-3-5	0.08	0.26	16.11	100.00
A1	<i>pn content</i> g/t	<i>pn content</i> g/t cum.	<i>pn distribution.</i> %	<i>pn distribution</i> % cum.
RC-1	1.17	1.17	66.02	66.02
RC-2	0.54	0.93	19.43	85.44
RC-3-5	0.20	0.61	14.56	100.00

Table 9 continued Pentlandite recovery to three flotation concentrates, assuming a 100% total recovery.

B1	<i>pn content g/t</i>	<i>pn content g/t cum.</i>	<i>pn distribution. %</i>	<i>pn distribution % cum.</i>
RC-1	1.23	1.23	58.72	58.72
RC-2	0.97	1.13	29.83	88.55
RC-3-5	0.18	0.70	11.45	100.00
A3	<i>pn content g/t</i>	<i>pn content g/t cum.</i>	<i>pn distribution. %</i>	<i>pn distribution % cum.</i>
RC-1	1.32	1.32	52.23	52.23
RC-2	0.83	1.08	31.72	83.95
RC-3-5	0.22	0.67	16.05	100.00
B3	<i>pn content g/t</i>	<i>pn content g/t cum.</i>	<i>pn distribution. %</i>	<i>pn distribution % cum.</i>
RC-1	2.12	2.12	67.11	67.11
RC-2	0.74	1.40	25.64	92.75
RC-3-5	0.10	0.71	7.25	100.00
A4	<i>pn content g/t</i>	<i>pn content g/t cum.</i>	<i>pn distribution. %</i>	<i>pn distribution % cum.</i>
RC-1	1.79	1.79	40.50	40.50
RC-2	1.52	1.65	38.85	79.35
RC-3-5	0.38	0.98	20.65	100.00
B4	<i>pn content g/t</i>	<i>pn content g/t cum.</i>	<i>pn distribution. %</i>	<i>pn distribution % cum.</i>
RC-1	0.63	0.63	53.88	53.88
RC-2	0.31	0.46	31.33	85.21
RC-3-5	0.05	0.21	14.79	100.00
A5	<i>pn content g/t</i>	<i>pn content g/t cum.</i>	<i>pn distribution. %</i>	<i>pn distribution % cum.</i>
RC-1	0.42	0.42	32.70	32.70
RC-2	0.45	0.44	35.85	68.55
RC-3-5	0.18	0.30	31.45	100.00
C1	<i>pn content g/t</i>	<i>pn content g/t cum.</i>	<i>pn distribution. %</i>	<i>pn distribution % cum.</i>
RC-1	0.04	0.04	57.92	57.92
RC-2	0.02	0.03	30.80	88.72
RC-3-5	0.00	0.01	11.28	100.00
C2	<i>pn content g/t</i>	<i>pn content g/t cum.</i>	<i>pn distribution. %</i>	<i>pn distribution % cum.</i>
RC-1	0.07	0.07	53.88	53.88
RC-2	0.03	0.05	25.85	79.72
RC-3-5	0.01	0.03	20.28	100.00
C3	<i>pn content g/t</i>	<i>pn content g/t cum.</i>	<i>pn distribution. %</i>	<i>pn distribution % cum.</i>
RC-1	0.09	0.09	48.53	48.53
RC-2	0.06	0.07	39.04	87.57
RC-3-5	0.01	0.04	12.43	100.00

Table 9 continued Pentlandite recovery to three flotation concentrates, assuming a 100% total recovery.

C4	<i>pn content g/t</i>	<i>pn content g/t cum.</i>	<i>pn distribution. %</i>	<i>pn distribution % cum.</i>
RC-1	0.07	0.07	67.38	67.38
RC-2	0.02	0.04	29.76	97.14
RC-3-5	0.00	0.02	2.86	100.00
C5	<i>pn content g/t</i>	<i>pn content g/t cum.</i>	<i>pn distribution. %</i>	<i>pn distribution % cum.</i>
RC-1	0.07	0.07	43.14	43.14
RC-2	0.06	0.07	45.59	88.73
RC-3-5	0.01	0.03	11.27	100.00
Comp.	<i>pn content g/t</i>	<i>pn content g/t cum.</i>	<i>pn distribution. %</i>	<i>pn distribution % cum.</i>
RC-1	0.77	0.77	53.31	53.31
RC-2	0.45	0.61	31.01	84.33
RC-3-5	0.11	0.35	15.67	100.00

Table 10 Pyrrhotite recovery to three flotation concentrates, assuming a 100% total recovery.

A2	<i>po content g/t</i>	<i>po content g/t cum.</i>	<i>po distribution. %</i>	<i>po distribution % cum.</i>
RC-1	0.24	0.24	41.80	41.80
RC-2	0.23	0.23	45.77	87.57
RC-3-5	0.04	0.14	12.43	100.00
B2	<i>po content g/t</i>	<i>po content g/t cum.</i>	<i>po distribution. %</i>	<i>po distribution % cum.</i>
RC-1	0.07	0.07	31.03	31.03
RC-2	0.10	0.09	60.71	91.74
RC-3-5	0.01	0.05	8.26	100.00
A1	<i>po content g/t</i>	<i>po content g/t cum.</i>	<i>po distribution. %</i>	<i>po distribution % cum.</i>
RC-1	0.48	0.48	28.97	28.97
RC-2	0.84	0.62	32.38	61.36
RC-3-5	0.50	0.57	38.64	100.00
B1	<i>po content g/t</i>	<i>po content g/t cum.</i>	<i>po distribution. %</i>	<i>po distribution % cum.</i>
RC-1	0.67	0.67	24.36	24.36
RC-2	1.71	1.08	40.20	64.56
RC-3-5	0.72	0.92	35.44	100.00
A3	<i>po content g/t</i>	<i>po content g/t cum.</i>	<i>po distribution. %</i>	<i>po distribution % cum.</i>
RC-1	0.36	0.36	14.27	14.27
RC-2	1.26	0.80	48.67	62.93
RC-3-5	0.51	0.66	37.07	100.00
B3	<i>po content g/t</i>	<i>po content g/t cum.</i>	<i>po distribution. %</i>	<i>po distribution % cum.</i>
RC-1	0.69	0.69	21.54	21.54
RC-2	1.49	1.11	51.05	72.58
RC-3-5	0.37	0.71	27.42	100.00

Table 10 continued Pyrrhotite recovery to three flotation concentrates, assuming a 100% total recovery.

A4	<i>po content</i> g/t	<i>po content</i> g/t cum.	<i>po distribution.</i> %	<i>po distribution</i> % cum.
RC-1	0.36	0.36	17.39	17.39
RC-2	0.66	0.52	35.37	52.76
RC-3-5	0.41	0.46	47.24	100.00
B4	<i>po content</i> g/t	<i>po content</i> g/t cum.	<i>po distribution.</i> %	<i>po distribution</i> % cum.
RC-1	0.50	0.50	47.12	47.12
RC-2	0.30	0.39	33.10	80.22
RC-3-5	0.06	0.19	19.78	100.00
A5	<i>po content</i> g/t	<i>po content</i> g/t cum.	<i>po distribution.</i> %	<i>po distribution</i> % cum.
RC-1	0.02	0.02	23.16	23.16
RC-2	0.03	0.02	45.02	68.18
RC-3-5	0.01	0.02	31.82	100.00
C1	<i>po content</i> g/t	<i>po content</i> g/t cum.	<i>po distribution.</i> %	<i>po distribution</i> % cum.
RC-1	0.01	0.01	17.41	17.41
RC-2	0.01	0.01	18.87	36.28
RC-3-5	0.02	0.02	63.72	100.00
C2	<i>po content</i> g/t	<i>po content</i> g/t cum.	<i>po distribution.</i> %	<i>po distribution</i> % cum.
RC-1	0.02	0.02	29.91	29.91
RC-2	0.02	0.02	36.55	66.46
RC-3-5	0.01	0.01	33.54	100.00
C3	<i>po content</i> g/t	<i>po content</i> g/t cum.	<i>po distribution.</i> %	<i>po distribution</i> % cum.
RC-1	0.01	0.01	11.95	11.95
RC-2	0.02	0.02	53.86	65.81
RC-3-5	0.01	0.01	34.19	100.00
C5	<i>po content</i> g/t	<i>po content</i> g/t cum.	<i>po distribution.</i> %	<i>po distribution</i> % cum.
RC-1	0.01	0.01	33.52	33.52
RC-2	0.01	0.01	66.48	100.00
RC-3-5	0.00	0.00	0.00	100.00
Comp.	<i>po content</i> g/t	<i>po content</i> g/t cum.	<i>po distribution.</i> %	<i>po distribution</i> % cum.
RC-1	0.26	0.26	21.80	21.80
RC-2	0.51	0.39	42.33	64.14
RC-3-5	0.20	0.29	35.86	100.00

Table 11 Pyrite recovery to three flotation concentrates, assuming a 100% total recovery.

A2	<i>py content</i> g/t	<i>py content</i> g/t cum.	<i>py distribution.</i> %	<i>py distribution</i> % cum.
RC-1	0.21	0.21	79.14	79.14
RC-2	0.04	0.12	16.87	96.01
RC-3-5	0.01	0.07	3.99	100.00

Table 11 continued Pyrite recovery to three flotation concentrates, assuming a 100% total recovery.

B2	<i>py content</i> g/t	<i>py content</i> g/t cum.	<i>py distribution.</i> %	<i>py distribution</i> % cum.
RC-1	0.51	0.51	66.72	66.72
RC-2	0.18	0.32	33.28	100.00
RC-3-5	0.00	0.16	0.00	100.00
A1	<i>py content</i> g/t	<i>py content</i> g/t cum.	<i>py distribution.</i> %	<i>py distribution</i> % cum.
RC-1	0.22	0.22	72.87	72.87
RC-2	0.05	0.15	10.38	83.25
RC-3-5	0.04	0.10	16.75	100.00
B1	<i>py content</i> g/t	<i>py content</i> g/t cum.	<i>py distribution.</i> %	<i>py distribution</i> % cum.
RC-1	0.00	0.00	7.71	7.71
RC-2	0.00	0.00	9.20	16.91
RC-3-5	0.01	0.01	83.09	100.00
A3	<i>py content</i> g/t	<i>py content</i> g/t cum.	<i>py distribution.</i> %	<i>py distribution</i> % cum.
RC-1	0.54	0.54	78.71	78.71
RC-2	0.10	0.32	14.92	93.63
RC-3-5	0.02	0.18	6.37	100.00
B3	<i>py content</i> g/t	<i>py content</i> g/t cum.	<i>py distribution.</i> %	<i>py distribution</i> % cum.
RC-1	0.05	0.05	51.28	51.28
RC-2	0.03	0.04	31.84	83.12
RC-3-5	0.01	0.02	16.88	100.00
A4	<i>py content</i> g/t	<i>py content</i> g/t cum.	<i>py distribution.</i> %	<i>py distribution</i> % cum.
RC-1	0.06	0.06	40.74	40.74
RC-2	0.02	0.04	13.56	54.30
RC-3-5	0.03	0.03	45.70	100.00
B4	<i>py content</i> g/t	<i>py content</i> g/t cum.	<i>py distribution.</i> %	<i>py distribution</i> % cum.
RC-1	0.04	0.04	89.52	89.52
RC-2	0.00	0.02	8.18	97.70
RC-3-5	0.00	0.01	2.30	100.00
A5	<i>py content</i> g/t	<i>py content</i> g/t cum.	<i>py distribution.</i> %	<i>py distribution</i> % cum.
RC-1	0.01	0.01	26.88	26.88
RC-2	0.02	0.01	71.34	98.22
RC-3-5	0.00	0.01	1.78	100.00
C1	<i>py content</i> g/t	<i>py content</i> g/t cum.	<i>py distribution.</i> %	<i>py distribution</i> % cum.
RC-1	0.23	0.23	58.45	58.45
RC-2	0.06	0.14	17.78	76.23
RC-3-5	0.03	0.08	23.77	100.00

Table 11 continued Pyrite recovery to three flotation concentrates, assuming a 100% total recovery

C2	<i>py content</i> g/t	<i>py content</i> g/t cum.	<i>py distribution.</i> %	<i>py distribution</i> % cum.
RC-1	1.04	1.04	74.84	74.84
RC-2	0.24	0.61	20.50	95.34
RC-3-5	0.02	0.28	4.66	100.00
C3	<i>py content</i> g/t	<i>py content</i> g/t cum.	<i>py distribution.</i> %	<i>py distribution</i> % cum.
RC-1	0.22	0.22	65.81	65.81
RC-2	0.09	0.15	32.29	98.10
RC-3-5	0.00	0.07	1.90	100.00
C4	<i>py content</i> g/t	<i>py content</i> g/t cum.	<i>py distribution.</i> %	<i>py distribution</i> % cum.
RC-1	0.67	0.67	76.95	76.95
RC-2	0.13	0.36	19.70	96.64
RC-3-5	0.01	0.18	3.36	100.00
C5	<i>py content</i> g/t	<i>py content</i> g/t cum.	<i>py distribution.</i> %	<i>py distribution</i> % cum.
RC-1	0.38	0.38	74.18	74.18
RC-2	0.09	0.21	21.80	95.97
RC-3-5	0.01	0.09	4.03	100.00
Comp	<i>py content</i> g/t	<i>py content</i> g/t cum.	<i>py distribution.</i> %	<i>py distribution</i> % cum.
RC-1	0.30	0.30	74.18	74.18
RC-2	0.08	0.19	18.68	92.86
RC-3-5	0.01	0.10	7.14	100.00

Table 12 Millerite recovery to three flotation concentrates, assuming a 100% total recovery

B2	<i>mil content</i> g/t	<i>mil content</i> g/t cum.	<i>mil distribution.</i> %	<i>mil distribution</i> % cum.
RC-1	0.05	0.05	48.59	48.59
RC-2	0.04	0.04	51.41	100.00
RC-3-5	0.00	0.02	0.00	100.00
C1	<i>mil content</i> g/t	<i>mil content</i> g/t cum.	<i>mil distribution.</i> %	<i>mil distribution</i> % cum.
RC-1	0.32	0.32	58.02	58.02
RC-2	0.12	0.21	25.84	83.86
RC-3-5	0.03	0.11	16.14	100.00
C2	<i>mil content</i> g/t	<i>mil content</i> g/t cum.	<i>mil distribution.</i> %	<i>mil distribution</i> % cum.
RC-1	0.48	0.48	60.21	60.21
RC-2	0.20	0.33	28.76	88.97
RC-3-5	0.03	0.16	11.03	100.00
C3	<i>mil content</i> g/t	<i>mil content</i> g/t cum.	<i>mil distribution.</i> %	<i>mil distribution</i> % cum.
RC-1	0.17	0.17	40.03	40.03
RC-2	0.08	0.12	20.72	60.75
RC-3-5	0.06	0.09	39.25	100.00



Table 12 continued *Millerite recovery to three flotation concentrates, assuming a 100% total recovery*

C4	<i>mil content g/t</i>	<i>mil content g/t cum.</i>	<i>mil distribution. %</i>	<i>mil distribution % cum.</i>
<i>RC-1</i>	0.24	0.24	66.92	66.92
<i>RC-2</i>	0.07	0.14	25.16	92.08
<i>RC-3-5</i>	0.01	0.07	7.92	100.00
C5	<i>mil content g/t</i>	<i>mil content g/t cum.</i>	<i>mil distribution. %</i>	<i>mil distribution % cum.</i>
<i>RC-1</i>	0.41	0.41	60.53	60.53
<i>RC-2</i>	0.17	0.28	31.63	92.16
<i>RC-3-5</i>	0.02	0.12	7.84	100.00
Comp.	<i>mil content g/t</i>	<i>mil content g/t cum.</i>	<i>mil distribution. %</i>	<i>mil distribution % cum.</i>
<i>RC-1</i>	0.28	0.28	62.83	62.83
<i>RC-2</i>	0.11	0.20	25.01	87.84
<i>RC-3-5</i>	0.03	0.11	12.16	100.00

Table 13 Relative proportions of **liberated** grains of different PGE minerals in flotation concentrates of samples A1, A3, A4, B4, A5, C1 and C2.

A1	RC1 (235)		RC2 (271)		RC3-5 (143)	
	% no.	area %	% no.	area %	% no.	area %
Pt,Pd,S	40	37	27	30	22	20
Pt,S	27	18	29	17	32	23
Pt,Rh,Cu,Ni,S	15	28	16	19	7	9
Ru,Os,Ir,S	9	7	13	22	13	21
PGE,Bi,Te	1	0	4	3	3	2
PGE,As,S	0	0	3	2	2	1
Pt,Pd,Rh,Fe	4	3	6	5	14	19
Pd,Pb,Sb,Hg	3	6	2	3	7	3
A3	RC1 (175)		RC2 (113)		RC3-5 (165)	
	% no.	area %	% no.	area %	% no.	area %
Pt,Pd,S	17	18	12	16	9	9
Pt,S	33	14	44	38	36	24
Pt,Rh,Cu,Ni,S	29	39	23	22	23	24
Ru,Os,Ir,S	10	15	7	16	12	18
PGE,Bi,Te	3	12	3	2	4	4
PGE,As,S	3	1	2	1	8	8
Pt,Pd,Rh,Fe	1	0	4	1	2	3
Pd,Pb,Sb,Hg	3	1	4	3	7	10
A4	RC1 (118)		RC2 (70)		RC3-5 (102)	
	% no.	area %	% no.	area %	% no.	area %
Pt,Pd,S	3	1	7	5	3	5
Pt,S	14	6	11	8	9	8
Pt,Rh,Cu,Ni,S	0	0	6	4	3	1
Ru,Os,Ir,S	16	42	17	29	17	30
PGE,Bi,Te	7	6	4	1	13	11
PGE,As,S	2	1	0	0	0	0
Pt,Pd,Rh,Fe	46	36	49	49	43	35
Pd,Pb,Sb,Hg	13	8	6	3	12	9
B4	RC1 (167)		RC2 (161)		RC3-5 (185)	
	% no.	area %	% no.	area %	% no.	area %
Pt,Pd,S	4	3	10	6	8	4
Pt,S	4	3	6	6	8	4
Pt,Rh,Cu,Ni,S	2	1	3	2	2	1
Ru,Os,Ir,S	25	36	25	40	22	41
PGE,Bi,Te	7	3	6	3	8	5
PGE,As,S	2	0	0	0	2	2
Pt,Pd,Rh,Fe	51	50	47	41	45	40
Pd,Pb,Sb,Hg	4	3	4	2	4	3
A5	RC1 (121)		RC2 (114)		RC3-5 (38)	
	% no.	area %	% no.	area %	% no.	area %
Pt,Pd,S	8	5	4	3	8	17
Pt,S	6	3	5	4	3	14
Pt,Rh,Cu,Ni,S	9	5	9	8	5	5
Ru,Os,Ir,S	20	39	16	34	24	29
PGE,Bi,Te	1	0	1	0	3	2
PGE,As,S	10	3	17	15	13	4
Pt,Pd,Rh,Fe	11	7	13	10	21	13
Pd,Pb,Sb,Hg	36	38	35	25	24	17

Table 13 continued Relative proportions of **liberated** grains of different PGE minerals in flotation concentrates of samples A1, A3, A4, B4, A5, C1 and C2.

C1	RC1 (235)		RC2 (118)		RC3-5 (134)	
	% no.	area %	% no.	area %	% no.	area %
<i>Pt,Pd,S</i>	47	63	38	37	33	26
<i>Pt,S</i>	5	6	9	4	7	4
<i>Pt,Rh,Cu,Ni,S</i>	28	17	36	38	31	35
<i>Ru,Os,Ir,S</i>	14	13	6	14	17	28
<i>PGE,Bi,Te</i>	1	0	3	1	2	1
<i>PGE,As,S</i>	1	0	5	3	6	3
<i>Pt,Pd,Rh,Fe</i>	0	0	1	1	0	0
<i>Pd,Pb,Sb,Hg</i>	3	1	3	1	4	3
C2	RC1 (229)		RC2 (230)		RC3-5 (169)	
	% no.	area %	% no.	area %	% no.	area %
<i>Pt,Pd,S</i>	41	48	36	31	30	23
<i>Pt,S</i>	7	5	5	1	5	5
<i>Pt,Rh,Cu,Ni,S</i>	39	38	44	45	35	26
<i>Ru,Os,Ir,S</i>	7	7	11	20	18	36
<i>PGE,Bi,Te</i>	1	1	1	0	0	0
<i>PGE,As,S</i>	2	0	4	2	10	9
<i>Pt,Pd,Rh,Fe</i>	0	0	0	0	0	0
<i>Pd,Pb,Sb,Hg</i>	4	1	0	0	3	0



APPENDIX I

Milling data

Table 1 *Milling data for fourteen samples of UG2 chromitite.*

Table 2 *Screen analysis of fourteen samples of UG2 chromitite milled to 80% <75 μ m.*

Table 1 Milling data, % Cr₂O₃ and median measured chromite equivalent circle diameter (ECD) of chromite for fourteen samples of UG2 chromitite.

Sample	A1		A2		A3		B1		B2		C1		C2	
	Time	% <75	Time	% <75	Time	% <75	Time	% <75	Time	% <75	Time	% <75	Time	% <75
Milling data	0	12.4	0	23.15	0	10.25	0	11.03	0	13.76	0	6.77	0	5.18
	36	36.41	30	42.61	30	34.92	35	47.35	30	55.94	30	35.41	30	50.65
	75	58.37	60	61.52	90	71.22	70	77.06	65	74.45	45	47.68	60	68.45
	120	81.14	95	80.35	105	79.55	75	82.34	72	79.5	61	81.7	72	80.29
	135	87.3												

Sample	C3		C4		C5		A4		A5		B3		B4	
	Time	% <75	Time	% <75	Time	% <75	Time	% <75	Time	% <75	Time	% <75	Time	% <75
Milling data	0	11.51	0	9.9	0	10.27	0	4.84	0	15.19	0	7.81	0	5.14
	30	65.59	30	35.11	30	44.96	60	65.48	30	71.91	40	50.05	60	70.77
	60	71.7	60	58.69	60	66.26	72	74.11	40	81	72	69.37	72	80.14
	70	80.61	95	79.25	75	80.76	80	80.02			82	75.22		
											90	80.29		

APPENDIX J

Flotation data

The following notation was used in all of the tables:

RC1=Rougher concentrate 1 (1 minute flotation)

RC2=Rougher concentrate 2 (3 minutes flotation)

RC3=Rougher concentrate 3 (8 minutes flotation)

RC4=Rougher concentrate 4 (15 minutes flotation)

RC5=Rougher concentrate 5 (20 minutes flotation)

RC3-5=combined rougher concentrates 3, 4 and 5 (8 to 20 minutes flotation)

RT=Rougher tailings

Table 1 *Mass recoveries for fourteen UG2 chromitite samples milled to 80% <75 μ m.*

Table 2 *Cr₂O₃ recoveries from fourteen UG2 chromitite samples milled to 80% <75 μ m.*

Table 3 *Silicate gangue recoveries from fourteen UG2 chromitite samples milled to 80% <75 μ m.*

Table 4 *Cu recoveries from fourteen test samples milled to 80% <75 μ m.*

Table 5 *Kelsall flotation constants for Cu recoveries from fourteen samples of UG2 chromitite milled to 80% <75 μ m.*

Table 6 *Acid soluble Ni recoveries from fourteen test samples milled to 80% <75 μ m.*

Table 7 *Kelsall flotation constants for acid soluble Ni recoveries from fourteen samples of UG2 chromitite milled to 80% <75 μ m.*

Table 8 *PGE+Au recoveries from fourteen samples of UG2 chromitite milled to 80% <75 μ m.*

Table 9 *Kelsall flotation constants for PGE+Au recoveries from fourteen samples of UG2 chromitite milled to 80% <75 μ m.*

Table 10 *Platinum recoveries from fourteen samples of UG2 chromitite milled to 80% <75 μ m.*

Table 11 *Kelsall flotation constants for platinum recoveries from fourteen samples of UG2 chromitite milled to 80% <75 μ m.*

Table 12 *Palladium recoveries from fourteen samples of UG2 chromitite milled to 80% <75 μ m.*

Table 13 *Kelsall flotation constants for palladium recoveries from fourteen samples of UG2 chromitite milled to 80% <75 μ m.*

Table 14 *Rhodium recoveries from fourteen samples of UG2 chromitite milled to 80% <75 μ m.*

Table 15 *Kelsall flotation constants for rhodium recoveries from fourteen samples of UG2 chromitite milled to 80% <75 μ m.*

Table 1 Mass recoveries for fourteen UG2 chromitite samples milled to 80% <75 μ m.

Confidence limits for cumulative recoveries were calculated at the 95% confidence level.

A2	0-1 min. RC1	1-3 min. RC2	3-8 min. RC3	8-15 min. RC4	15-20 min. RC5	RT
Dry (g)	158.5	180.1	171.1	84.9	29.6	5367.6
Cum. Dry (g)	158.5	338.6	509.7	594.6	624.2	5991.8
Dry (%)	2.6	3.0	2.9	1.4	0.5	89.6
Cum. Dry (%)	2.6 \pm 0.3	5.7 \pm 0.4	8.5 \pm 0.5	9.9 \pm 0.5	10.4 \pm 0.5	100.0
B2	0-1 min. RC1	1-3 min. RC2	3-8 min. RC3	8-15 min. RC4	15-20 min. RC5	RT
Dry (g)	160.1	225.9	229.1	106.7	38.4	5241.3
Cum. Dry (g)	160.1	386.0	615.1	721.8	760.2	6001.5
Dry (%)	2.7	3.8	3.8	1.8	0.6	87.3
Cum. Dry (%)	2.7 \pm 0.3	6.4 \pm 0.6	10.2 \pm 0.7	12.0 \pm 0.7	12.7 \pm 0.8	100.0
A1	0-1 min. RC1	1-3 min. RC2	3-8 min. RC3	8-15 min. RC4	15-20 min. RC5	RT
Dry (g)	673.4	431.8	487.2	281.1	103.5	16004.9
Cum. Dry (g)	673.4	1105.2	1592.4	1873.5	1977.0	17981.9
Dry (%)	3.7	2.4	2.7	1.6	0.6	89.0
Cum. Dry (%)	3.7 \pm 0.3	6.1 \pm 0.4	8.9 \pm 0.5	10.4 \pm 0.5	11.0 \pm 0.5	100.0
B1	0-1 min. RC1	1-3 min. RC2	3-8 min. RC3	8-15 min. RC4	15-20 min. RC5	RT
Dry (g)	218.0	141.2	163.1	97.3	32.5	5339.6
Cum. Dry (g)	218.0	359.2	522.3	619.6	652.1	5991.7
Dry (%)	3.6	2.4	2.7	1.6	0.5	89.1
Cum. Dry (%)	3.6 \pm 0.4	6.0 \pm 0.6	8.7 \pm 0.7	10.3 \pm 0.8	10.9 \pm 0.8	100.0
A3	0-1 min. RC1	1-3 min. RC2	3-8 min. RC3	8-15 min. RC4	15-20 min. RC5	RT
Dry (g)	179.7	174.0	192.1	100.8	37.0	5309.6
Cum. Dry (g)	179.7	353.7	545.8	646.6	683.6	5993.2
Dry (%)	3.0	2.9	3.2	1.7	0.6	88.6
Cum. Dry (%)	3.0 \pm 0.6	5.9 \pm 0.8	9.1 \pm 0.9	10.8 \pm 1.0	11.4 \pm 1.0	100.0
B3	0-1 min. RC1	1-3 min. RC2	3-8 min. RC3	8-15 min. RC4	15-20 min. RC5	RT
Dry (g)	139.0	165.2	229.7	164.9	66.5	5215.2
Cum. Dry (g)	139.0	304.2	533.9	698.8	765.3	5980.5
Dry (%)	2.3	2.8	3.8	2.8	1.1	87.2
Cum. Dry (%)	2.3 \pm 0.1	5.1 \pm 0.2	8.9 \pm 0.2	11.7 \pm 0.2	12.8 \pm 0.2	100.0

Table 1 cont. Mass recoveries for fourteen UG2 chromitite samples milled to 80% <75µm. .

Confidence limits for cumulative recoveries were calculated at the 95% confidence level.

A4	0-1 min. RC1	1-3 min. RC2	3-8 min. RC3	8-15 min. RC4	15-20 min. RC5	RT
Dry (g)	135.3	152.7	179.2	105.0	40.4	5362.1
Cum. Dry (g)	135.3	288.0	467.2	572.2	612.6	5974.7
Dry (%)	2.3	2.6	3.0	1.8	0.7	89.7
Cum. Dry (%)	2.3± 0.1	4.8± 0.2	7.8± 0.2	9.6± 0.4	10.3± 0.4	100.0
B4	0-1 min. RC1	1-3 min. RC2	3-8 min. RC3	8-15 min. RC4	15-20 min. RC5	RT
Dry (g)	141.7	154.4	175.7	112.6	47.5	5356.1
Cum. Dry (g)	141.7	296.1	471.8	584.4	631.9	5988.0
Dry (%)	2.4	2.6	2.9	1.9	0.8	89.4
Cum. Dry (%)	2.4± 0.3	4.9± 0.6	7.9± 0.8	9.8± 0.8	10.6± 0.7	100.0
A5	0-1 min. RC1	1-3 min. RC2	3-8 min. RC3	8-15 min. RC4	15-20 min. RC5	RT
Dry (g)	173.5	179.0	202.9	127.2	53.3	5242.0
Cum. Dry (g)	173.5	352.5	555.4	682.6	735.9	5977.9
Dry (%)	2.9	3.0	3.4	2.1	0.9	87.7
Cum. Dry (%)	2.9± 0.2	5.9± 0.2	9.3± 0.2	11.4± 0.3	12.3± 0.3	100.0
C1	0-1 min. RC1	1-3 min. RC2	3-8 min. RC3	8-15 min. RC4	15-20 min. RC5	RT
Dry (g)	137.9	166.0	209.0	135.1	50.0	5281.8
Cum. Dry (g)	137.9	303.9	512.9	648.0	698.0	5979.8
Dry (%)	2.3	2.8	3.5	2.3	0.8	88.3
Cum. Dry (%)	2.3± 0.2	5.1± 0.3	8.6± 0.3	10.8± 0.4	11.7± 0.3	100.0± 0.2
C2	0-1 min. RC1	1-3 min. RC2	3-8 min. RC3	8-15 min. RC4	15-20 min. RC5	RT
Dry (g)	170.7	200.6	243.8	169.7	70.6	5215.9
Cum. Dry (g)	170.7	371.3	615.1	784.8	855.4	6071.3
Dry (%)	2.8	3.3	4.0	2.8	1.2	85.9
Cum. Dry (%)	2.8± 0.1	6.1± 0.3	10.1± 0.3	12.9± 0.3	14.1± 0.3	100.0
C3	0-1 min. RC1	1-3 min. RC2	3-8 min. RC3	8-15 min. RC4	15-20 min. RC5	RT
Dry (g)	146.8	168.5	195.4	133.9	61.9	5319.8
Cum. Dry (g)	146.8	315.3	510.7	644.6	706.5	6026.3
Dry (%)	2.4	2.8	3.2	2.2	1.0	88.3
Cum. Dry (%)	2.4± 0.2	5.2± 0.3	8.5± 0.5	10.7± 0.5	11.7± 0.6	100.0

Table 1 cont. Mass recoveries for fourteen UG2 chromitite samples milled to 80% <75 μ m.
Confidence limits for cumulative recoveries were calculated at the 95% confidence level.

C4	0-1 min. RC1	1-3 min. RC2	3-8 min. RC3	8-15 min. RC4	15-20 min. RC5	RT
Dry (g)	180.4	237.0	259.3	148.1	57.6	5108.5
Cum. Dry (g)	180.4	417.4	676.7	824.8	882.4	5990.9
Dry (%)	3.0	4.0	4.3	2.5	1.0	85.3
Cum. Dry (%)	3.0 \pm 0.1	7.0 \pm 0.1	11.3 \pm 0.3	13.8 \pm 0.4	14.7 \pm 0.5	100.0
C5	0-1 min. RC1	1-3 min. RC2	3-8 min. RC3	8-15 min. RC4	15-20 min. RC5	RT
Dry (g)	119.1	151.5	201.8	137.4	58.1	5318.3
Cum. Dry (g)	119.1	270.6	472.4	609.8	667.9	5986.2
Dry (%)	2.0	2.5	3.4	2.3	1.0	88.8
Cum. Dry (%)	2.0 \pm 0.1	4.5 \pm 0.2	7.9 \pm 0.2	10.2 \pm 0.2	11.2 \pm 0.3	100.0

Table 2 Cr_2O_3 recovery from fourteen UG2 chromitite samples milled to 80% <75 μ m.

A2	Cr₂O₃ content %	Cr₂O₃ content cum. %	Cr₂O₃ distr. %	Cr₂O₃ distr. cum. %	Chromite mass %
RC1	18.7	18.7	1.7	1.7	40
RC2	19.4	19.1	2.0	3.7	41
RC3	19.3	19.2	1.9	5.6	41
RC4	19.5	19.2	1.0	6.5	41
RC5	19.3	19.2	0.3	6.8	41
RT	30.4	29.2	93.2	100.0	65

Cr_2O_3 calculated = 29.2% Cr_2O_3 assay = 29.1% Difference = 0.1% (%Difference = 0%)

B2	Cr₂O₃ content %	Cr₂O₃ content cum. %	Cr₂O₃ distr. %	Cr₂O₃ distr. cum. %	Chromite mass %
RC1	12.2	12.2	1.6	1.6	26
RC2	12.8	12.6	2.3	3.9	27
RC3	12.8	12.6	2.4	6.3	27
RC4	12.9	12.7	1.1	7.4	27
RC5	12.6	12.7	0.4	7.8	27
RT	21.8	20.6	92.2	100.0	46

Cr_2O_3 calculated = 20.6% Cr_2O_3 assay = 20.9% Difference = 0.3% (%Difference = 1%)

A1	Cr₂O₃ content %	Cr₂O₃ content cum. %	Cr₂O₃ distr. %	Cr₂O₃ distr. cum. %	Chromite mass %
RC1	21.9	21.9	2.8	2.8	47
RC2	21.8	21.9	1.8	4.5	46
RC3	22.6	22.1	2.1	6.6	48
RC4	22.6	22.2	1.2	7.8	48
RC5	22.4	22.2	0.4	8.2	48
RT	30.5	29.6	91.8	100.0	65

Cr_2O_3 calculated = 29.6% Cr_2O_3 assay = 29.3% Difference = 0.6% (%Difference = 1%)

B1	Cr₂O₃ content %	Cr₂O₃ content cum. %	Cr₂O₃ distr. %	Cr₂O₃ distr. cum. %	Chromite mass %
RC1	13.2	13.2	1.9	1.9	28
RC2	14.3	13.6	1.3	3.2	30
RC3	14.5	13.9	1.5	4.7	31
RC4	14.6	14.0	0.9	5.6	31
RC5	14.5	14.0	0.3	5.9	31
RT	27.2	25.8	94.1	100.0	58

Cr_2O_3 calculated = 25.8% Cr_2O_3 assay = 25.1% Difference = 0.7% (%Difference = 3%)

Table 2 cont. Cr_2O_3 recovery from fourteen UG2 chromitite samples milled to 80% < 75 μ m.

A3	Cr₂O₃ content %	Cr₂O₃ content cum. %	Cr₂O₃ distr. %	Cr₂O₃ distr. cum. %	Chromite mass %
RC1	24.9	24.9	2.2	2.2	53
RC2	25.8	25.3	2.2	4.4	55
RC3	25.6	25.4	2.4	6.8	54
RC4	25.8	25.5	1.3	8.1	55
RC5	25.5	25.5	0.5	8.6	54
RT	35.0	33.9	91.4	100.0	74

Cr_2O_3 calculated = 33.9% Cr_2O_3 assay = 33.8% Difference = 0.1% (%Difference = 0%)

B3	Cr₂O₃ content %	Cr₂O₃ content cum. %	Cr₂O₃ distr. %	Cr₂O₃ distr. cum. %	Chromite mass %
RC-1	20.1	20.1	1.5	1.5	43
RC-2	20.4	20.3	1.7	3.3	43
RC-3	20.6	20.4	2.0	5.2	44
RC-4	20.1	20.3	1.2	6.4	43
RC-5	20.4	20.3	0.5	7.0	43
RT	32.1	30.9	93.1	100.0	68

Cr_2O_3 calculated = 30.9% Cr_2O_3 assay = 30.7% Difference = 0.2% (%Difference = 1%)

A4	Cr₂O₃ content %	Cr₂O₃ content cum. %	Cr₂O₃ distr. %	Cr₂O₃ distr. cum. %	Chromite mass %
RC1	25.1	25.1	1.6	1.6	53
RC2	25.4	25.3	1.9	3.5	54
RC3	25.8	25.5	2.2	5.7	55
RC4	26.0	25.6	1.3	7.0	55
RC5	26.7	25.6	0.5	7.5	57
RT	35.9	34.9	92.5	100.0	76

Cr_2O_3 calculated = 34.9% Cr_2O_3 assay = 34.5% Difference = 0.4% (%Difference = 1%)

B4	Cr₂O₃ content %	Cr₂O₃ content cum. %	Cr₂O₃ distr. %	Cr₂O₃ distr. cum. %	Chromite mass %
RC1	23.0	23.0	1.6	1.6	49
RC2	24.2	23.7	2.0	3.7	51
RC3	24.4	24.0	2.8	6.5	52
RC4	24.4	24.1	2.0	8.5	52
RC5	24.8	24.1	0.8	9.4	53
RT	34.3	33.0	90.6	100.0	73

Cr_2O_3 calculated = 33.0% Cr_2O_3 assay = 32.9% Difference = 0.1% (%Difference = 0%)

A5	Cr₂O₃ content %	Cr₂O₃ content cum. %	Cr₂O₃ distr. %	Cr₂O₃ distr. cum. %	Chromite mass %
RC1	16.9	16.9	1.8	1.8	36
RC2	17.8	17.4	1.9	3.7	38
RC3	18.7	17.9	2.3	5.9	40
RC4	19.3	18.1	1.5	7.4	41
RC5	20.0	18.3	0.6	8.0	43
RT	29.5	28.1	92.0	100.0	63

Cr_2O_3 calculated = 28.1% Cr_2O_3 assay = 27.7% Difference = 0.4% (%Difference = 1%)

Table 2 cont. Cr_2O_3 recovery from fourteen UG2 chromitite samples milled to 80% <75 μ m.

C1	Cr₂O₃ content %	Cr₂O₃ content cum. %	Cr₂O₃ distr. %	Cr₂O₃ distr. cum. %	Chromite mass %
RC1	15.8	15.8	1.1	1.1	34
RC2	17.6	16.8	1.4	2.5	37
RC3	19.4	17.9	2.0	4.4	41
RC4	21.4	18.6	1.4	5.8	45
RC5	22.0	18.8	0.5	6.4	47
RT	36.7	34.6	93.7	100.0	78

Cr_2O_3 calculated = 34.6% Cr_2O_3 assay = 34.6% Difference = 0.0% (%Difference = 0%)

C2	Cr₂O₃ content %	Cr₂O₃ content cum. %	Cr₂O₃ distr. %	Cr₂O₃ distr. cum. %	Chromite mass %
RC1	16.3	16.3	1.5	1.5	35
RC2	17.1	16.7	1.8	3.3	36
RC3	17.9	17.2	2.3	5.6	38
RC4	19.1	17.6	1.7	7.3	41
RC5	19.3	17.7	0.7	8.0	41
RT	33.5	31.3	92.0	100.0	71

Cr_2O_3 calculated = 31.3% Cr_2O_3 assay = 31.8% Difference = 0.5% (%Difference = 2%)

C3	Cr₂O₃ content %	Cr₂O₃ content cum. %	Cr₂O₃ distr. %	Cr₂O₃ distr. cum. %	Chromite mass %
RC1	14.1	14.1	1.1	1.1	30
RC2	14.8	14.5	1.2	2.3	31
RC3	15.8	15.0	1.6	3.9	34
RC4	17.3	15.5	1.2	5.1	37
RC5	17.8	15.7	0.6	5.7	38
RT	34.7	32.5	94.4	100.0	74

Cr_2O_3 calculated = 32.5% Cr_2O_3 assay = 32.4% Difference = 0.1% (%Difference = 0%)

C4	Cr₂O₃ content %	Cr₂O₃ content cum. %	Cr₂O₃ distr. %	Cr₂O₃ distr. cum. %	Chromite mass %
RC1	17.2	17.2	2.0	2.0	37
RC2	17.7	17.4	1.3	3.3	38
RC3	18.2	17.7	1.6	4.9	39
RC4	19.0	17.9	1.0	5.8	40
RC5	18.8	17.9	0.3	6.2	40
RT	33.3	31.6	93.8	100.0	71

Cr_2O_3 calculated = 31.6% Cr_2O_3 assay = 31.2% Difference = 0.4% (%Difference = 1%)

C5	Cr₂O₃ content %	Cr₂O₃ content cum. %	Cr₂O₃ distr. %	Cr₂O₃ distr. cum. %	Chromite mass %
RC1	20.7	20.7	1.2	1.2	44
RC2	21.6	21.2	1.6	2.8	46
RC3	22.0	21.5	2.2	5.0	47
RC4	22.8	21.8	1.6	6.6	48
RC5	22.8	21.9	0.7	7.3	48
RT	35.2	33.7	92.8	100.0	75

Cr_2O_3 calculated = 33.7% Cr_2O_3 assay = 33.8% Difference = 0.1% (%Difference = 0%)

Table 3 Silicate gangue recoveries from fourteen UG2 chromitite samples at a grind of 80% <75 μ m.

A1	<i>Silicate content</i> g	<i>Silicate content</i> cum. g	<i>Silicate distr.</i> %	<i>Silicate distr.</i> cum. %
RC1	119.9	119.9	5.4	5.4
RC2	77.3	197.2	3.5	8.9
RC3	84.4	281.6	3.8	12.7
RC4	48.7	330.3	2.2	14.9
RC5	18.1	348.4	0.8	15.7
RT	1873.0	2221.4	84.3	100.0

A2	<i>Silicate content</i> g	<i>Silicate content</i> cum. g	<i>Silicate distr.</i> %	<i>Silicate distr.</i> cum. %
RC1	95.5	95.5	4.2	4.2
RC2	105.8	201.3	4.7	8.9
RC3	100.9	302.2	4.4	13.3
RC4	49.7	352.0	2.2	15.5
RC5	17.5	369.4	0.8	16.3
RT	1899.3	2268.7	83.7	100.0

B1	<i>Silicate content</i> g	<i>Silicate content</i> cum. g	<i>Silicate distr.</i> %	<i>Silicate distr.</i> cum. %
RC1	156.8	156.8	5.8	5.8
RC2	98.3	255.1	3.6	9.4
RC3	112.8	368.0	4.2	13.6
RC4	67.1	435.1	2.5	16.1
RC5	22.5	457.5	0.8	16.9
RT	2252.6	2710.1	83.1	100.0

B2	<i>Silicate content</i> g	<i>Silicate content</i> cum. g	<i>Silicate distr.</i> %	<i>Silicate distr.</i> cum. %
RC1	118.6	118.6	3.5	3.5
RC2	164.4	283.0	4.9	8.4
RC3	166.8	449.8	5.0	13.4
RC4	77.4	527.2	2.3	15.7
RC5	28.1	555.4	0.8	16.5
RT	2812.7	3368.0	83.5	100.0

A3	<i>Silicate content</i> g	<i>Silicate content</i> cum. g	<i>Silicate distr.</i> %	<i>Silicate distr.</i> cum. %
RC1	84.6	84.6	5.1	5.1
RC2	78.6	163.2	4.7	9.8
RC3	87.6	250.7	5.2	15.0
RC4	45.5	296.3	2.7	17.7
RC5	16.9	313.2	1.0	18.7
RT	1359.6	1672.9	81.3	100.0

Table 3 cont. Silicate gangue recoveries from fourteen UG2 chromitite samples at a grind of 80% < 75µm.

A4	<i>Silicate content</i> g	<i>Silicate content</i> cum. g	<i>Silicate distr.</i> %	<i>Silicate distr.</i> cum. %
RC1	63.1	63.1	4.1	4.1
RC2	70.3	133.4	4.5	8.6
RC3	80.9	214.3	5.2	13.8
RC4	47.0	261.3	3.0	16.9
RC5	17.5	278.8	1.1	18.0
RT	1270.5	1549.3	82.0	100.0

B3	<i>Silicate content</i> g	<i>Silicate content</i> cum. g	<i>Silicate distr.</i> %	<i>Silicate distr.</i> cum. %
RC1	81.2	81.2	3.9	3.9
RC2	87.5	168.6	4.2	8.2
RC3	98.8	267.4	4.8	13.0
RC4	64.5	331.9	3.1	16.1
RC5	26.9	358.8	1.3	17.4
RT	1701.7	2060.5	82.6	100.0

B4	<i>Silicate content</i> g	<i>Silicate content</i> cum. g	<i>Silicate distr.</i> %	<i>Silicate distr.</i> cum. %
RC1	71.0	71.0	4.0	4.0
RC2	80.2	151.3	4.5	8.5
RC3	110.6	261.8	6.2	14.7
RC4	79.6	341.4	4.5	19.1
RC5	31.4	372.8	1.8	20.9
RT	1413.1	1785.9	79.1	100.0

A5	<i>Silicate content</i> g	<i>Silicate content</i> cum. g	<i>Silicate distr.</i> %	<i>Silicate distr.</i> cum. %
RC1	111.2	111.2	4.6	4.6
RC2	111.3	222.5	4.6	9.2
RC3	122.3	344.7	5.1	14.3
RC4	75.0	419.7	3.1	17.4
RC5	30.6	450.4	1.3	18.7
RT	1960.7	2411.1	81.3	100.0

C1	<i>Silicate content</i> g	<i>Silicate content</i> cum. g	<i>Silicate distr.</i> %	<i>Silicate distr.</i> cum. %
RC1	91.6	91.6	5.8	5.8
RC2	103.9	195.5	6.6	12.4
RC3	122.8	318.3	7.8	20.1
RC4	73.6	392.0	4.7	24.8
RC5	26.6	418.6	1.7	26.5
RT	1161.7	1580.3	73.5	100.0

Table 3 cont. Silicate gangue recoveries from fourteen UG2 chromitite samples at a grind of 80% <75 μ m.

C2	<i>Silicate content</i> g	<i>Silicate content</i> cum. g	<i>Silicate distr.</i> %	<i>Silicate distr.</i> cum. %
RC1	111.6	111.6	5.5	5.5
RC2	127.7	239.2	6.3	11.8
RC3	151.3	390.6	7.4	19.2
RC4	100.8	491.4	5.0	24.1
RC5	41.6	533.0	2.0	26.2
RT	1501.9	2034.9	73.8	100.0

C3	<i>Silicate content</i> g	<i>Silicate content</i> cum. g	<i>Silicate distr.</i> %	<i>Silicate distr.</i> cum. %
RC1	102.8	102.8	5.5	5.5
RC2	115.5	218.3	6.2	11.7
RC3	129.8	348.1	6.9	18.6
RC4	84.8	432.9	4.5	23.2
RC5	38.5	471.4	2.1	25.2
RT	1396.2	1867.5	74.8	100.0

C4	<i>Silicate content</i> g	<i>Silicate content</i> cum. g	<i>Silicate distr.</i> %	<i>Silicate distr.</i> cum. %
RC1	138.3	138.3	7.0	7.0
RC2	88.1	226.4	4.5	11.5
RC3	100.0	326.4	5.1	16.6
RC4	58.0	384.4	3.0	19.6
RC5	19.5	403.9	1.0	20.6
RT	1560.3	1964.2	79.4	100.0

C5	<i>Silicate content</i> g	<i>Silicate content</i> cum. g	<i>Silicate distr.</i> %	<i>Silicate distr.</i> cum. %
RC1	66.7	66.7	3.9	3.9
RC2	81.9	148.6	4.8	8.8
RC3	107.4	256.1	6.3	15.1
RC4	70.8	326.9	4.2	19.3
RC5	29.9	356.8	1.8	21.0
RT	1339.3	1696.1	79.0	100.0

Table 4 Cu recoveries from fourteen test samples. Differences between calculated and assayed Cu values for feed material are expressed as relative %.

A2	Cu content %	Cu content cum. %	Cu distribution %	Cu distribution cum. %
RC1	0.130	0.130	56.88	56.88
RC2	0.028	0.076	13.92	70.80
RC3	0.018	0.056	8.50	79.30
RC4	0.016	0.051	3.75	83.05
RC5	0.017	0.049	1.39	84.44
RT	0.001	0.006	15.56	100.00

Cu calculated = 0.006% *Cu assay* = 0.005% %Difference = 21%

B2	Cu content %	Cu content cum. %	Cu distribution %	Cu distribution cum. %
RC1	0.110	0.110	57.85	57.85
RC2	0.015	0.054	11.13	68.98
RC3	0.012	0.039	9.03	78.02
RC4	0.010	0.034	3.51	81.52
RC5	0.010	0.033	1.26	82.78
RT	0.001	0.005	17.22	100.00

Cu calculated = 0.001% *Cu assay* = 0.004% %Difference = 27%

A1	Cu content %	Cu content cum. %	Cu distribution %	Cu distribution cum. %
RC1	0.120	0.12	62.21	62.21
RC2	0.036	0.09	11.88	74.09
RC3	0.020	0.07	7.17	81.26
RC4	0.018	0.06	3.65	84.91
RC5	0.014	0.06	1.16	86.08
RT	0.001	0.007	13.92	100.00

Cu calculated = 0.007% *Cu assay* = 0.007% %Difference = 5%

B1	Cu content %	Cu content cum. %	Cu distribution %	Cu distribution cum. %
RC1	0.220	0.220	75.68	75.68
RC2	0.031	0.146	6.91	82.59
RC3	0.020	0.106	5.15	87.73
RC4	0.019	0.093	2.92	90.65
RC5	0.018	0.089	0.92	91.57
RT	0.001	0.011	8.43	100.00

Cu calculated = 0.011% *Cu assay* = 0.010% %Difference = 6%

A3	Cu content %	Cu content cum. %	Cu distribution %	Cu distribution cum. %
RC1	0.220	0.220	55.79	55.79
RC2	0.058	0.140	14.24	70.03
RC3	0.034	0.103	9.22	79.25
RC4	0.031	0.092	4.41	83.66
RC5	0.026	0.088	1.36	85.01
RT	0.002	0.012	14.99	100.00

Cu calculated = 0.012% *Cu assay* = 0.011% %Difference = 7%

Table 4 cont. Cu recoveries from fourteen test samples. Differences between calculated and assayed Cu values for feed material are expressed as relative %.

B3	<i>Cu content</i> %	<i>Cu content</i> <i>cum. %</i>	<i>Cu distribution</i> %	<i>Cu distribution</i> <i>cum. %</i>
RC1	0.279	0.279	67.53	67.53
RC2	0.050	0.160	13.19	80.72
RC3	0.020	0.108	6.00	86.72
RC4	0.016	0.090	3.08	89.80
RC5	0.013	0.084	1.05	90.85
RT	0.001	0.010	9.15	100.00

Cu calculated = 0.010% *Cu assay* = 0.009% %Difference = 9%

A4	<i>Cu content</i> %	<i>Cu content</i> <i>cum. %</i>	<i>Cu distribution</i> %	<i>Cu distribution</i> <i>cum. %</i>
RC1	0.266	0.266	50.85	50.85
RC2	0.082	0.168	17.69	68.55
RC3	0.043	0.120	10.89	79.44
RC4	0.028	0.103	4.15	83.59
RC5	0.022	0.098	1.26	84.85
RT	0.002	0.012	15.15	100.00

Cu calculated = 0.012% *Cu assay* = 0.011% %Difference = 8%

B4	<i>Cu content</i> %	<i>Cu content</i> <i>cum. %</i>	<i>Cu distribution</i> %	<i>Cu distribution</i> <i>cum. %</i>
RC1	0.457	0.457	63.86	63.86
RC2	0.083	0.254	13.80	77.66
RC3	0.041	0.162	9.51	87.17
RC4	0.021	0.129	3.47	90.64
RC5	0.018	0.119	1.23	91.87
RT	0.002	0.017	8.13	100.00

Cu calculated = 0.017% *Cu assay* = 0.017% %Difference = 2%

A5	<i>Cu content</i> %	<i>Cu content</i> <i>cum. %</i>	<i>Cu distribution</i> %	<i>Cu distribution</i> <i>cum. %</i>
RC1	0.090	0.090	52.36	52.36
RC2	0.022	0.055	13.20	65.57
RC3	0.014	0.040	9.53	75.09
RC4	0.013	0.035	5.54	80.64
RC5	0.010	0.033	1.79	82.42
RT	0.001	0.005	17.58	100.00

Cu calculated = 0.005% *Cu assay* = 0.004% %Difference = 25%

C1	<i>Cu content</i> %	<i>Cu content</i> <i>cum. %</i>	<i>Cu distribution</i> %	<i>Cu distribution</i> <i>cum. %</i>
RC1	0.095	0.095	31.48	31.48
RC2	0.027	0.058	10.86	42.34
RC3	0.017	0.041	8.40	50.74
RC4	0.015	0.035	4.74	55.49
RC5	0.015	0.034	1.81	57.30
RT	0.003	0.007	42.70	100.00

Cu calculated = 0.007% *Cu assay* = 0.006% %Difference = 19%

Table 4 cont. *Cu recoveries from fourteen test samples. Differences between calculated and assayed Cu values for feed material are expressed as relative %.*

C2	<i>Cu content</i> %	<i>Cu content</i> <i>cum. %</i>	<i>Cu distribution</i> %	<i>Cu distribution</i> <i>cum. %</i>
<i>RC1</i>	0.076	0.076	34.16	34.16
<i>RC2</i>	0.023	0.047	11.95	46.11
<i>RC3</i>	0.015	0.034	9.36	55.47
<i>RC4</i>	0.013	0.029	5.80	61.26
<i>RC5</i>	0.011	0.028	2.15	63.41
<i>RT</i>	0.003	0.006	36.59	100.00

Cu calculated = 0.006% *Cu assay* = 0.004% %Difference = 40%

C3	<i>Cu content</i> %	<i>Cu content</i> <i>cum. %</i>	<i>Cu distribution</i> %	<i>Cu distribution</i> <i>cum. %</i>
<i>RC1</i>	0.068	0.068	28.30	28.30
<i>RC2</i>	0.023	0.044	11.16	39.46
<i>RC3</i>	0.015	0.033	8.46	47.91
<i>RC4</i>	0.013	0.029	5.11	53.02
<i>RC5</i>	0.013	0.027	2.29	55.31
<i>RT</i>	0.003	0.006	44.69	100.00

Cu calculated = 0.006% *Cu assay* = 0.005% %Difference = 17%

C4	<i>Cu content</i> %	<i>Cu content</i> <i>cum. %</i>	<i>Cu distribution</i> %	<i>Cu distribution</i> <i>cum. %</i>
<i>RC1</i>	0.041	0.041	30.59	30.59
<i>RC2</i>	0.014	0.026	13.91	44.51
<i>RC3</i>	0.010	0.020	10.45	54.96
<i>RC4</i>	0.009	0.018	5.51	60.47
<i>RC5</i>	0.008	0.017	1.91	62.38
<i>RT</i>	0.002	0.004	37.62	100.00

Cu calculated = 0.004% *Cu assay* = 0.003% %Difference = 38%

C5	<i>Cu content</i> %	<i>Cu content</i> <i>cum. %</i>	<i>Cu distribution</i> %	<i>Cu distribution</i> <i>cum. %</i>
<i>RC1</i>	0.088	0.088	38.68	38.68
<i>RC2</i>	0.025	0.053	13.80	52.48
<i>RC3</i>	0.016	0.037	12.14	64.63
<i>RC4</i>	0.011	0.031	5.79	70.42
<i>RC5</i>	0.010	0.030	2.18	72.60
<i>RT</i>	0.001	0.005	27.40	100.00

Cu calculated = 0.005% *Cu assay* = 0.003% % Difference = 32%

Table 5 Kelsall flotation constants for Cu recoveries from fourteen samples of UG2 chromitite milled to 80% <75 μ m.

Cu	<i>U</i>	ϕ	K_f	K_s	<i>Loss</i>
A2	85.6	0.28	2.02	0.18	0.10
B2	83.1	0.31	2.06	0.2	0.12
A1	86.8	0.24	2.24	0.16	0.02
B1	92.3	0.17	3.08	0.15	0.00
A3	85.8	0.30	2.01	0.17	0.03
B3	91.5	0.19	2.07	0.16	0.02
A4	85.2	0.36	1.73	0.21	0.08
B4	92.6	0.24	2.05	0.14	0.01
A5	84.0	0.34	2.04	0.14	0.01
C1	58.7	0.42	1.76	0.14	0.05
C2	65.5	0.43	1.74	0.13	0.03
C3	57.5	0.44	1.57	0.12	0.11
C4	63.7	0.46	1.49	0.15	0.05
C5	73.6	0.47	1.92	0.17	0.19
<i>Avg.</i>	79.0	0.33	1.98	0.16	0.06

$$Loss = (Observed - Predicted)^2$$

Table 6 Acid soluble Ni recoveries from fourteen test samples. Differences between calculated and assayed Ni values for feed material are expressed as relative %.

A2	Ni content %	Ni content cum. %	Ni distribution %	Ni distribution cum. %
RC1	0.320	0.320	28.86	28.86
RC2	0.140	0.224	14.35	43.21
RC3	0.108	0.185	10.52	53.73
RC4	0.101	0.173	4.88	58.61
RC5	0.100	0.170	1.68	60.29
RT	0.013	0.029	39.71	100.00

Ni calculated = 0.029% Ni assay = 0.029% %Difference = 1%

B2	Ni content %	Ni content cum. %	Ni distribution %	Ni distribution cum. %
RC1	0.280	0.280	24.56	24.56
RC2	0.089	0.168	11.03	35.60
RC3	0.077	0.134	9.72	45.31
RC4	0.073	0.125	4.29	49.61
RC5	0.075	0.123	1.57	51.18
RT	0.017	0.030	48.82	100.00

Ni calculated = 0.030% Ni assay = 0.030% %Difference = 1%

A1	Ni content %	Ni content cum. %	Ni distribution %	Ni distribution cum. %
RC1	0.392	0.392	35.66	35.66
RC2	0.175	0.307	10.16	45.82
RC3	0.120	0.251	7.77	53.59
RC4	0.105	0.229	3.84	57.43
RC5	0.146	0.225	2.07	59.50
RT	0.019	0.042	40.50	100.00

Ni calculated = 0.042% Ni assay = 0.041% %Difference = 2%

B1	Ni content %	Ni content cum. %	Ni distribution %	Ni distribution cum. %
RC1	0.405	0.405	21.82	21.82
RC2	0.209	0.328	7.29	29.11
RC3	0.159	0.275	6.39	35.50
RC4	0.160	0.257	3.84	39.33
RC5	0.161	0.252	1.29	40.62
RT	0.045	0.068	59.38	100.00

Ni calculated = 0.068% Ni assay = 0.072% %Difference = 6%

A3	Ni content %	Ni content cum. %	Ni distribution %	Ni distribution cum. %
RC1	0.450	0.450	30.24	30.24
RC2	0.200	0.327	12.98	43.23
RC3	0.145	0.263	10.42	53.65
RC4	0.130	0.242	4.90	58.55
RC5	0.126	0.236	1.74	60.28
RT	0.020	0.045	39.72	100.00

Ni calculated = 0.045% Ni assay = 0.044% %Difference = 1%

Table 6 cont. Acid soluble Ni recoveries from fourteen test samples. Differences between calculated and assayed Ni values for feed material are expressed as relative %.

A4	<i>Ni content</i> %	<i>Ni content</i> cum. %	<i>Ni distribution</i> %	<i>Ni distribution</i> cum. %
RC1	0.648	0.648	25.36	25.36
RC2	0.370	0.501	16.34	41.71
RC3	0.253	0.406	13.12	54.82
RC4	0.199	0.368	6.04	60.87
RC5	0.163	0.354	1.90	62.77
RT	0.024	0.058	37.23	100.00

Ni calculated = 0.058% *Ni assay* = 0.055% %Difference = 5%

A5	<i>Ni content</i> %	<i>Ni content</i> cum. %	<i>Ni distribution</i> %	<i>Ni distribution</i> cum. %
RC1	0.232	0.232	17.85	17.85
RC2	0.133	0.182	10.56	28.40
RC3	0.109	0.155	9.81	38.21
RC4	0.107	0.146	6.04	44.25
RC5	0.097	0.143	2.29	46.54
RT	0.023	0.038	53.46	100.00

Ni calculated = 0.038% *Ni assay* = 0.035% %Difference = 8%

B3	<i>Ni content</i> %	<i>Ni content</i> cum. %	<i>Ni distribution</i> %	<i>Ni distribution</i> cum. %
RC1	0.612	0.612	35.25	35.25
RC2	0.190	0.392	11.92	47.17
RC3	0.130	0.294	9.28	56.45
RC4	0.105	0.258	4.81	61.26
RC5	0.090	0.245	1.74	62.99
RT	0.017	0.041	37.01	100.00

Ni calculated = 0.041% *Ni assay* = 0.041% %Difference = 0%

B4	<i>Ni content</i> %	<i>Ni content</i> cum. %	<i>Ni distribution</i> %	<i>Ni distribution</i> cum. %
RC1	0.738	0.738	31.58	31.58
RC2	0.279	0.489	14.20	45.78
RC3	0.211	0.369	14.89	60.67
RC4	0.145	0.316	7.34	68.01
RC5	0.119	0.299	2.43	70.44
RT	0.018	0.054	29.56	100.00

Ni calculated = 0.054% *Ni assay* = 0.056% %Difference = 3%

Table 6 cont. Acid soluble Ni recoveries from fourteen test samples. Differences between calculated and assayed Ni values for feed material are expressed as relative %.

C1	<i>Ni content</i> %	<i>Ni content</i> cum. %	<i>Ni distribution</i> %	<i>Ni distribution</i> cum. %
<i>RC1</i>	0.201	0.201	17.11	17.11
<i>RC2</i>	0.099	0.145	10.12	27.23
<i>RC3</i>	0.080	0.118	10.34	37.58
<i>RC4</i>	0.078	0.110	6.53	44.10
<i>RC5</i>	0.079	0.108	2.45	46.55
<i>RT</i>	0.016	0.027	53.45	100.00

Ni calculated = 0.027% *Ni assay* = 0.028% %Difference = 4%

C2	<i>Ni content</i> %	<i>Ni content</i> cum. %	<i>Ni distribution</i> %	<i>Ni distribution</i> cum. %
<i>RC1</i>	0.228	0.228	19.35	19.35
<i>RC2</i>	0.107	0.163	10.67	30.02
<i>RC3</i>	0.093	0.135	11.21	41.23
<i>RC4</i>	0.089	0.125	7.51	48.73
<i>RC5</i>	0.083	0.121	2.91	51.65
<i>RT</i>	0.019	0.033	48.35	100.00

Ni calculated = 0.033% *Ni assay* = 0.030% %Difference = 10%

C3	<i>Ni content</i> %	<i>Ni content</i> cum. %	<i>Ni distribution</i> %	<i>Ni distribution</i> cum. %
<i>RC1</i>	0.150	0.150	12.62	12.62
<i>RC2</i>	0.090	0.118	8.69	21.31
<i>RC3</i>	0.083	0.105	9.29	30.60
<i>RC4</i>	0.083	0.100	6.37	36.97
<i>RC5</i>	0.084	0.099	2.98	39.95
<i>RT</i>	0.020	0.029	60.05	100.00

Ni calculated = 0.029% *Ni assay* = 0.029% %Difference = 0%

C4	<i>Ni content</i> %	<i>Ni content</i> cum. %	<i>Ni distribution</i> %	<i>Ni distribution</i> cum. %
<i>RC1</i>	0.138	0.138	15.14	15.14
<i>RC2</i>	0.083	0.107	11.97	27.11
<i>RC3</i>	0.071	0.093	11.22	38.34
<i>RC4</i>	0.070	0.089	6.35	44.68
<i>RC5</i>	0.071	0.088	2.49	47.17
<i>RT</i>	0.017	0.027	52.83	100.00

Ni calculated = 0.027% *Ni assay* = 0.026% %Difference = 6%

C5	<i>Ni content</i> %	<i>Ni content</i> cum. %	<i>Ni distribution</i> %	<i>Ni distribution</i> cum. %
<i>RC1</i>	0.231	0.231	16.32	16.32
<i>RC2</i>	0.111	0.164	9.93	26.25
<i>RC3</i>	0.081	0.128	9.66	35.92
<i>RC4</i>	0.080	0.117	6.49	42.40
<i>RC5</i>	0.078	0.114	2.69	45.10
<i>RT</i>	0.017	0.028	54.90	100.00

Ni calculated = 0.028% *Ni assay* = 0.028% %Difference = 1%

Table 7 Kelsall flotation constants for acid soluble Ni recoveries from fourteen samples of UG2 chromitite milled to 80% <75 μ m.

Ni (a.s.)	U	ϕ	K_f	K_s	Loss
A1	60.6	0.4	1.9	0.1	0.3
A2	61.2	0.5	1.4	0.2	0.1
A3	61.1	0.5	1.6	0.2	0.1
B1	41.8	0.5	1.9	0.1	0.0
B2	51.8	0.5	1.7	0.2	0.1
C1	49.1	0.6	1.4	0.1	0.0
C2	55.0	0.6	1.5	0.1	0.0
C3	43.9	0.7	1.2	0.1	0.1
C4	49.5	0.6	1.0	0.1	0.1
C5	48.4	0.6	1.3	0.1	0.0
A4	63.8	0.6	1.2	0.2	0.1
A5	48.8	0.6	1.3	0.1	0.0
B3	64.1	0.4	1.8	0.2	0.1
B4	68.5	0.7	1.4	0.2	0.2
Avg.	54.8	0.6	1.5	0.1	0.1

Table 8 PGE+Au recoveries from fourteen samples of UG2 chromitite milled to 80% <75 μ m.

A2	PGE+Au content ppm	PGE+Au content cumulative ppm	PGE+Au dist. %	PGE+Au dist. cumulative %
RC1	95.60	95.60	69.47	69.47
RC2	16.87	53.72	13.93	83.39
RC3	8.62	38.58	6.76	90.16
RC4	6.87	34.05	2.67	92.83
RC5	5.70	32.71	0.77	93.60
RT	0.26	3.64	6.40	100.00

PGE+Au calculated = 3.64 ppm PGE+Au assay = 3.08 ppm Difference = 0.56 ppm
(%Difference = 18%)

B2	PGE+Au content ppm	PGE+Au content cumulative ppm	PGE+Au dist. %	PGE+Au dist. cumulative %
RC1	110.90	110.90	78.82	78.82
RC2	9.02	51.28	9.05	87.87
RC3	5.33	34.16	5.42	93.29
RC4	3.74	29.66	1.77	95.06
RC5	3.06	28.32	0.52	95.58
RT	0.19	3.75	4.42	100.00

PGE+Au calculated = 3.75 ppm PGE+Au assay = 3.55 ppm Difference = 0.20 ppm
(%Difference = 6%)

A1	PGE+Au content ppm	PGE+Au content cumulative ppm	PGE+Au dist. %	PGE+Au dist. cumulative %
RC1	103.18	103.18	75.22	75.22
RC2	25.66	72.90	11.99	87.21
RC3	11.26	54.05	5.94	93.15
RC4	7.20	47.02	2.19	95.34
RC5	5.79	44.86	0.65	95.99
RT	0.23	5.14	4.01	100.00

PGE+Au calculated = 5.14 ppm PGE+Au assay = 5.15 ppm Difference = 0.01 ppm
(%Difference = 0%)

B1	PGE+Au content ppm	PGE+Au content cumulative ppm	PGE+Au dist. %	PGE+Au dist. cumulative %
RC1	66.61	66.61	70.75	70.75
RC2	16.33	46.84	11.23	81.98
RC3	8.64	34.91	6.86	88.84
RC4	6.65	30.47	3.15	91.99
RC5	6.23	29.26	0.99	92.98
RT	0.27	3.43	7.02	100.00

PGE+Au calculated = 3.43 ppm PGE+Au assay = 4.10 ppm Difference = 0.67 ppm
(%Difference = 16%)

Table 8 cont. PGE+Au recoveries from fourteen samples of UG2 chromitite milled to 80% <75 μ m.

A3	PGE+Au content ppm	PGE+Au content cumulative ppm	PGE+Au dist. %	PGE+Au dist. cumulative %
RC1	125.70	125.70	60.34	60.34
RC2	32.74	79.97	15.22	75.56
RC3	17.60	58.02	9.03	84.59
RC4	13.35	51.05	3.59	88.19
RC5	7.61	48.70	0.75	88.94
RT	0.78	6.25	11.06	100.00

PGE+Au calculated = 6.25 ppm PGE+Au assay = 5.85 ppm Difference = 0.40 ppm
(%Difference = 7%)

A4	PGE+Au content ppm	PGE+Au content cumulative ppm	PGE+Au dist. %	PGE+Au dist. cumulative %
RC1	135.05	135.05	54.72	54.72
RC2	46.63	88.16	21.32	76.05
RC3	22.55	63.00	12.10	88.15
RC4	13.05	53.83	4.10	92.25
RC5	9.58	50.91	1.16	93.41
RT	0.410	5.59	6.59	100.00

PGE+Au calculated = 5.59 ppm PGE+Au assay = 5.02 ppm Difference = 0.57 ppm
(%Difference = 11%)

B3	PGE+Au content ppm	PGE+Au content cumulative ppm	PGE+Au dist. %	PGE+Au dist. cumulative %
RC1	156.71	156.71	69.17	69.17
RC2	33.00	92.20	15.87	85.04
RC3	13.60	62.93	7.44	92.49
RC4	6.40	52.04	2.24	94.73
RC5	3.93	48.42	0.58	95.31
RT	0.28	5.36	4.69	100.00

PGE+Au calculated = 5.36 ppm PGE+Au assay = 4.70 ppm Difference = 0.66 ppm
(%Difference = 14%)

B4	PGE+Au content ppm	PGE+Au content cumulative ppm	PGE+Au dist. %	PGE+Au dist. cumulative %
RC1	148.77	148.77	57.98	57.98
RC2	38.02	88.62	17.61	75.58
RC3	20.25	59.21	13.04	88.62
RC4	9.205	47.41	4.26	92.88
RC5	6.27	43.83	1.17	94.05
RT	0.41	5.96	5.95	100.00

PGE+Au calculated = 5.96 ppm PGE+Au assay = 5.81 ppm Difference = 0.15 ppm
(%Difference = 3%)

Table 8 cont. PGE+Au recovery from fourteen samples of UG2 chromitite milled to 80% <75 μ m.

A5	<i>PGE+Au content ppm</i>	<i>PGE+Au content cumulative ppm</i>	<i>PGE+Au dist. %</i>	<i>PGE+Au dist. cumulative %</i>
RC1	71.27	71.27	35.72	35.72
RC2	26.99	48.78	13.95	49.67
RC3	17.30	37.28	10.14	59.81
RC4	14.55	33.05	5.35	65.16
RC5	12.55	31.56	1.93	67.09
RT	2.17	5.79	32.91	100.00

*PGE+Au calculated = 5.79 ppm PGE+Au assay = 5.74 ppm Difference = 0.05 ppm
(%Difference = 1%)*

C1	<i>PGE+Au content ppm</i>	<i>PGE+Au content cumulative ppm</i>	<i>PGE+Au dist. %</i>	<i>PGE+Au dist. cumulative %</i>
RC1	135.01	135.01	61.95	61.95
RC2	28.35	76.75	15.66	77.61
RC3	12.35	50.51	8.59	86.19
RC4	7.56	41.55	3.40	89.59
RC5	5.52	38.97	0.92	90.51
RT	0.54	5.03	9.49	100.00

*PGE+Au calculated = 5.03 ppm PGE+Au assay = 4.39 ppm Difference = 0.64 ppm
(%Difference = 14%)*

C2	<i>PGE+Au content ppm</i>	<i>PGE+Au content cumulative ppm</i>	<i>PGE+Au dist. %</i>	<i>PGE+Au dist. cumulative %</i>
RC1	119.30	119.30	66.49	66.49
RC2	22.25	66.87	14.57	81.07
RC3	10.15	44.39	8.08	89.15
RC4	6.24	36.14	3.46	92.60
RC5	4.39	33.52	1.01	93.61
RT	0.375	5.04	6.39	100.00

*PGE+Au calculated = 5.04 ppm PGE+Au assay = 4.58 ppm Difference = 0.46 ppm
(%Difference = 10%)*

C3	<i>PGE+Au content ppm</i>	<i>PGE+Au content cumulative ppm</i>	<i>PGE+Au dist. %</i>	<i>PGE+Au dist. cumulative %</i>
RC1	152.66	152.66	63.34	63.34
RC2	32.05	88.20	15.26	78.60
RC3	15.45	60.37	8.53	87.14
RC4	9.09	49.72	3.44	90.58
RC5	7.03	45.98	1.23	91.81
RT	0.545	5.87	8.19	100.00

*PGE+Au calculated = 5.87 ppm PGE+Au assay = 4.92 ppm Difference = 0.95 ppm
(%Difference = 19%)*

Table 8 cont. PGE+Au recoveries from fourteen samples of UG2 chromitite milled to 80% <75 μ m.

C4	PGE+Au content ppm	PGE+Au content cumulative ppm	PGE+Au dist. %	PGE+Au dist. cumulative %
RC1	110.59	110.59	70.83	70.83
RC2	16.08	56.93	13.53	84.36
RC3	7.60	38.03	7.00	91.36
RC4	4.84	32.07	2.54	93.90
RC5	3.20	30.18	0.65	94.56
RT	0.300	4.70	5.44	100.00

PGE+Au calculated = 4.70 ppm PGE+Au assay = 4.31 ppm Difference = 0.39 ppm
(%Difference = 9%)

C5	PGE+Au content ppm	PGE+Au content cumulative ppm	PGE+Au dist. %	PGE+Au dist. cumulative %
RC1	176.89	176.89	69.83	69.83
RC2	28.80	93.98	14.46	84.29
RC3	11.35	58.68	7.59	91.88
RC4	6.44	46.91	2.93	94.81
RC5	4.26	43.20	0.82	95.63
RT	0.248	5.04	4.37	100.00

PGE+Au calculated = 5.04 ppm PGE+Au assay = 4.86 ppm Difference = 0.18 ppm
(%Difference = 4%)

Table 9 Kelsall flotation constants for PGE+Au recoveries from fourteen samples of UG2 chromitite milled to 80% <75 μ m.

PGE+Au	U	ϕ	K_f	K_s	Loss
A2	93.85	0.20	2.13	0.20	0.02
B2	95.63	0.16	2.90	0.24	0.02
A1	96.13	0.17	2.38	0.22	0.02
B1	93.44	0.21	2.45	0.18	0.02
A3	89.29	0.29	2.05	0.21	0.00
B3	95.36	0.22	2.06	0.25	0.03
A4	93.63	0.36	1.60	0.22	0.08
B4	94.19	0.40	2.17	0.24	0.10
A5	68.40	0.42	1.62	0.15	0.07
C1	90.83	0.27	1.98	0.21	0.02
C2	94.02	0.24	2.10	0.19	0.03
C3	92.15	0.26	2.01	0.19	0.10
C4	94.73	0.21	2.24	0.22	0.02
C5	95.87	0.22	2.16	0.21	0.03
Avg.	91.97	0.26	2.13	0.21	0.04

$$\text{Loss} = (\text{Observed} - \text{Predicted})^2$$

Table 10 Platinum recovery from fourteen samples of UG2 chromitite milled to 80% <75 μ m.

A2	<i>Pt content ppm</i>	<i>Pt content cumulative ppm</i>	<i>Pt distribution %</i>	<i>Pt distribution cumulative %</i>
RC1	64.13	64.13	70.87	70.87
RC2	10.90	35.82	13.69	84.56
RC3-5	5.13	21.77	10.21	94.76
RT	0.14	2.39	5.24	100.00

Pt calculated = 2.39 ppm *Pt assay* = 2.13 ppm *Difference* = 0.26 ppm
Difference = 11%

B2	<i>Pt content ppm</i>	<i>Pt content cumulative ppm</i>	<i>Pt distribution %</i>	<i>Pt distribution cumulative %</i>
RC1	65.87	65.87	79.49	79.49
RC2	5.39	30.48	9.18	88.67
RC3-5	2.81	16.85	7.91	96.58
RT	0.09	2.21	3.42	100.00

Pt calculated = 2.21 ppm *Pt assay* = 2.04 ppm *Difference* = 0.17 ppm
Difference = 8%

A1	<i>Pt content ppm</i>	<i>Pt content cumulative ppm</i>	<i>Pt distribution %</i>	<i>Pt distribution cumulative %</i>
RC1	61.83	61.83	74.97	74.97
RC2	15.48	43.73	12.03	87.00
RC3-5	5.75	26.98	9.01	96.02
RT	0.14	3.09	3.98	100.00

Pt calculated = 3.09 ppm *Pt assay* = 3.08 ppm *Difference* = 0.18 ppm
Difference = 0%

B1	<i>Pt content ppm</i>	<i>Pt content cumulative ppm</i>	<i>Pt distribution %</i>	<i>Pt distribution cumulative %</i>
RC1	39.63	39.63	71.63	71.63
RC2	8.47	27.38	9.92	81.54
RC3-5	4.32	17.02	10.49	92.03
RT	0.18	2.01	7.97	100.00

Pt calculated = 2.01 ppm *Pt assay* = 2.41 ppm *Difference* = 0.40 ppm
Difference = 20%

A3	<i>Pt content ppm</i>	<i>Pt content cumulative ppm</i>	<i>Pt distribution %</i>	<i>Pt distribution cumulative %</i>
RC1	80.85	80.85	63.31	63.31
RC2	19.8	50.82	15.01	78.32
RC3-5	9.13	30.70	13.12	91.44
RT	0.37	3.83	8.56	100.00

Pt calculated = 3.83 ppm *Pt assay* = 3.67 ppm *Difference* = 0.16 ppm
Difference = 4%

Table 10 cont. Platinum recoveries from fourteen samples of UG2 chromitite milled to 80% <math><75\mu\text{m}</math>.

B3	<i>Pt content ppm</i>	<i>Pt content cumulative ppm</i>	<i>Pt distribution %</i>	<i>Pt distribution cumulative %</i>
RC1	94.50	94.50	69.75	69.75
RC2	20.00	55.65	16.08	85.83
RC3-5	5.39	28.94	9.43	95.26
RT	0.17	3.21	4.74	100.00

*Pt calculated = 3.21 ppm Pt assay = 2.81 ppm Difference = 0.40 ppm
Difference = 13%*

A4	<i>Pt content ppm</i>	<i>Pt content cumulative ppm</i>	<i>Pt distribution %</i>	<i>Pt distribution cumulative %</i>
RC1	88.50	88.50	59.09	59.09
RC2	27.00	55.89	20.35	79.44
RC3-5	9.20	31.15	14.74	94.18
RT	0.22	3.39	5.82	100.00

*Pt calculated = 3.39 ppm Pt assay = 3.05 ppm Difference = 0.34 ppm
% Difference = 10%*

B4	<i>Pt content ppm</i>	<i>Pt content cumulative ppm</i>	<i>Pt distribution %</i>	<i>Pt distribution cumulative %</i>
RC1	92.50	92.50	62.39	62.39
RC2	21.00	53.67	16.83	79.22
RC3-5	6.91	25.50	15.46	94.69
RT	0.21	3.45	5.31	100.00

*Pt calculated = 3.45 ppm Pt assay = 3.34 ppm Difference = 0.11 ppm
Difference = 3%*

A5	<i>Pt content ppm</i>	<i>Pt content cumulative ppm</i>	<i>Pt distribution %</i>	<i>Pt distribution cumulative %</i>
RC1	39.50	39.50	36.63	36.63
RC2	15.10	27.11	14.45	51.08
RC3-5	8.15	17.23	16.70	67.78
RT	1.15	3.13	32.22	100.00

*Pt calculated = 3.13 ppm Pt assay = 3.13 ppm Difference = 0 ppm
Difference = 0 %*

C1	<i>Pt content ppm</i>	<i>Pt content cumulative ppm</i>	<i>Pt distribution %</i>	<i>Pt distribution cumulative %</i>
RC1	92.00	92.00	62.67	62.67
RC2	19.50	52.40	15.99	78.66
RC3-5	6.68	26.58	12.99	91.65
RT	0.32	3.39	8.35	100.00

*Pt calculated = 3.39 ppm Pt assay = 3.03 ppm Difference = 0.36 ppm
Difference = 11%*

Table 10 cont. Platinum recovery from fourteen samples of UG2 chromitite milled to 80% <75 μ m.

C2	<i>Pt content ppm</i>	<i>Pt content cumulative ppm</i>	<i>Pt distribution %</i>	<i>Pt distribution cumulative %</i>
RC1	75.50	75.50	65.76	65.76
RC2	14.70	42.65	15.05	80.81
RC3-5	5.40	21.57	13.34	94.14
RT	0.22	3.23	5.86	100.00

Pt calculated = 3.23 ppm Pt assay = 2.97 ppm Difference = 0.26 ppm
Difference = 8%

C3	<i>Pt content ppm</i>	<i>Pt content cumulative ppm</i>	<i>Pt distribution %</i>	<i>Pt distribution cumulative %</i>
RC1	102.25	102.25	65.42	65.42
RC2	20.50	58.56	15.06	80.48
RC3-5	7.30	30.18	12.45	92.93
RT	0.31	3.81	7.07	100.00

Pt calculated = 3.81 ppm Pt assay = 3.25 ppm Difference = 0.56 ppm
Difference = 15%

C4	<i>Pt content ppm</i>	<i>Pt content cumulative ppm</i>	<i>Pt distribution %</i>	<i>Pt distribution cumulative %</i>
RC1	75.25	75.25	70.68	70.68
RC2	10.85	38.68	13.39	84.06
RC3-5	4.55	20.70	11.02	95.08
RT	0.19	3.21	4.92	100.00

Pt calculated = 3.21 ppm Pt assay = 2.90 ppm Difference = 0.31 ppm
Difference = 10%

C5	<i>Pt content ppm</i>	<i>Pt content cumulative ppm</i>	<i>Pt distribution %</i>	<i>Pt distribution cumulative %</i>
RC1	113.50	113.50	70.55	70.55
RC2	18.60	60.37	14.71	85.26
RC3-5	5.30	27.61	10.99	96.25
RT	0.14	3.20	3.75	100.00

Pt calculated = 3.20 ppm Pt assay = 3.06 ppm Difference = 0.14 ppm
Difference = 4%



Table 11 Kelsall flotation constants for platinum recoveries from fourteen samples of UG2 chromitite milled to 80% <75µm assuming an ultimate recovery of 100 per cent.

Pt	ϕ	K_f	K_s	<i>Loss</i>
A2	0.18	1.91	0.06	0.00
B2	0.14	2.44	0.07	0.00
A1	0.15	2.13	0.07	0.00
B1	0.21	2.27	0.05	0.00
A3	0.25	1.74	0.05	0.00
B3	0.17	1.74	0.06	0.00
A4	0.24	1.41	0.07	0.00
B4	0.26	1.68	0.08	0.00
A5	0.52	1.31	0.02	0.00
C1	0.25	1.68	0.05	0.00
C2	0.23	1.81	0.07	0.00
C3	0.23	1.78	0.06	0.00
C4	0.19	1.97	0.07	0.00
C5	0.18	1.88	0.08	0.00
Avg.	0.23	1.84	0.06	0.00

$$Loss = (Observed - Predicted)^2$$

Table 12 Palladium recoveries from fourteen samples of UG2 chromitite milled to 80% <75 μ m.

A2	<i>Pd content ppm</i>	<i>Pd content cumulative ppm</i>	<i>Pd distribution %</i>	<i>Pd distribution cumulative %</i>
RC1	17.03	17.03	65.07	65.07
RC2	3.70	9.94	16.06	81.14
RC3-5	1.80	6.22	12.39	93.53
RT	0.05	0.69	6.47	100.00

Pd calculated = 0.69 ppm *Pd assay* = 0.66 ppm *Difference* = 0.03 ppm
Difference = 6%

B2	<i>Pd content ppm</i>	<i>Pd content cumulative ppm</i>	<i>Pd distribution %</i>	<i>Pd distribution cumulative %</i>
RC1	33.67	33.67	77.12	77.12
RC2	2.79	15.60	9.02	86.13
RC3-5	1.68	8.75	8.99	95.13
RT	0.07	1.16	4.87	100.00

Pd calculated = 1.16 ppm *Pd assay* = 1.17 ppm *Difference* = 0.01 ppm
Difference = 0%

A1	<i>Pd content ppm</i>	<i>Pd content cumulative ppm</i>	<i>Pd distribution %</i>	<i>Pd distribution cumulative %</i>
RC1	29.27	29.27	73.82	73.82
RC2	7.21	20.65	11.66	85.48
RC3-5	2.89	12.82	9.43	94.91
RT	0.09	1.49	5.09	100.00

Pd calculated = 1.49 ppm *Pd assay* = 1.42 ppm *Difference* = 0.07 ppm
Difference = 5%

B1	<i>Pd content ppm</i>	<i>Pd content cumulative ppm</i>	<i>Pd distribution %</i>	<i>Pd distribution cumulative %</i>
RC1	20.03	20.03	66.33	66.33
RC2	5.79	14.43	12.42	78.75
RC3-5	2.77	9.19	12.33	91.08
RT	0.11	1.10	8.92	100.00

Pd calculated = 1.10 ppm *Pd assay* = 1.30 ppm *Difference* = 0.20 ppm
Difference = 15%

A3	<i>Pd content ppm</i>	<i>Pd content cumulative ppm</i>	<i>Pd distribution %</i>	<i>Pd distribution cumulative %</i>
RC1	30.95	30.95	56.45	56.45
RC2	9.05	20.18	15.98	72.44
RC3-5	4.90	12.80	16.41	88.84
RT	0.21	1.64	11.16	100.00

Pd calculated = 1.64 ppm *Pd assay* = 1.57 ppm *Difference* = 0.07 ppm
Difference = 5%

Table 12 cont. Palladium recoveries from fourteen samples of UG2 chromitite milled to 80% < 75µm.

B3	<i>Pd content</i> ppm	<i>Pd content</i> cumulative ppm	<i>Pd distribution</i> %	<i>Pd distribution</i> cumulative %
RC1	45.50	45.50	69.16	69.16
RC2	9.15	26.55	15.15	84.31
RC3-5	2.92	13.99	10.52	94.83
RT	0.09	1.56	5.17	100.00

Pd calculated = 1.56 ppm *Pd assay* = 1.43 ppm *Difference* = 0.13 ppm
Difference = 8%

A4	<i>Pd content</i> ppm	<i>Pd content</i> cumulative ppm	<i>Pd distribution</i> %	<i>Pd distribution</i> cumulative %
RC1	33.50	33.50	47.71	47.71
RC2	13.90	23.11	22.34	70.06
RC3-5	6.29	14.19	21.48	91.53
RT	0.15	1.59	8.47	100.00

Pd calculated = 1.59 ppm *Pd assay* = 1.48 ppm *Difference* = 0.11 ppm
Difference = 7%

B4	<i>Pd content</i> ppm	<i>Pd content</i> cumulative ppm	<i>Pd distribution</i> %	<i>Pd distribution</i> cumulative %
RC1	42.00	42.00	50.93	50.93
RC2	12.65	26.06	18.23	69.16
RC3-5	5.97	13.96	24.01	93.18
RT	0.15	1.92	6.82	100.00

Pd calculated = 1.92 ppm *Pd assay* = 1.87 ppm *Difference* = 0.05 ppm
Difference = 2%

A5	<i>Pd content</i> ppm	<i>Pd content</i> cumulative ppm	<i>Pd distribution</i> %	<i>Pd distribution</i> cumulative %
RC1	26.00	26.00	37.06	37.06
RC2	9.15	17.44	13.46	50.52
RC3-5	5.05	10.98	15.89	66.41
RT	0.78	2.04	33.59	100.00

Pd calculated = 2.04 ppm *Pd assay* = 2.05 ppm *Difference* = 0.01 ppm
Difference = 0%

C1	<i>Pd content</i> ppm	<i>Pd content</i> cumulative ppm	<i>Pd distribution</i> %	<i>Pd distribution</i> cumulative %
RC1	31.00	31.00	63.71	63.71
RC2	5.35	16.99	13.24	76.95
RC3-5	2.12	8.59	12.42	89.37
RT	0.135	1.12	10.63	100.00

Pd calculated = 1.12 ppm *Pd assay* = 1.04 ppm *Difference* = 0.08 ppm
Difference = 7%

C2	<i>Pd content</i> ppm	<i>Pd content</i> cumulative ppm	<i>Pd distribution</i> %	<i>Pd distribution</i> cumulative %
RC1	32.50	32.50	69.90	69.90
RC2	4.70	17.48	11.88	81.78
RC3-5	1.86	8.64	11.32	93.10
RT	0.105	1.31	6.90	100.00

Pd calculated = 1.31 ppm *Pd assay* = 1.14 ppm *Difference* = 0.17 ppm



Difference = 13%

C3	<i>Pd content ppm</i>	<i>Pd content cumulative ppm</i>	<i>Pd distribution %</i>	<i>Pd distribution cumulative %</i>
<i>RC1</i>	36.00	36.00	63.33	63.33
<i>RC2</i>	7.08	20.54	14.29	77.61
<i>RC3-5</i>	2.60	10.61	12.19	89.80
<i>RT</i>	0.16	1.38	10.20	100.00

Pd calculated = 1.38 ppm Pd assay = 1.12 ppm Difference = 0.26 ppm
Difference = 19%



Table 12 cont. Palladium recoveries from fourteen samples of UG2 chromitite milled to 80% <75µm.

C4	<i>Pd content ppm</i>	<i>Pd content cumulative ppm</i>	<i>Pd distribution %</i>	<i>Pd distribution cumulative %</i>
RC1	23.75	23.75	70.12	70.12
RC2	3.20	12.08	12.41	82.53
RC3-5	1.34	6.42	10.20	92.73
RT	0.09	1.02	7.27	100.00

*Pd calculated = 1.02 ppm Pd assay = 0.94 ppm Difference = 0.08 ppm
Difference = 8%*

C5	<i>Pd content ppm</i>	<i>Pd content cumulative ppm</i>	<i>Pd distribution %</i>	<i>Pd distribution cumulative %</i>
RC1	47.50	47.50	71.01	71.01
RC2	6.65	24.63	12.65	83.66
RC3-5	2.17	11.27	10.80	94.46
RT	0.08	1.33	5.54	100.00

*Pd calculated = 1.33 ppm Pd assay = 1.31ppm Difference = 0.02 ppm
Difference = 2%*

Table 13 Kelsall flotation constants for Pd recoveries from fourteen samples of UG2 chromitite milled to 80% <75µm assuming an ultimate recovery of 100%.

Pd	ϕ	K_f	K_s	<i>Loss</i>
A2	0.22	1.71	0.06	0.00
B2	0.17	2.44	0.06	0.00
A1	0.17	2.11	0.06	0.00
B1	0.24	1.98	0.05	0.00
A3	0.31	1.65	0.05	0.00
B3	0.19	1.80	0.06	0.00
A4	0.35	1.19	0.07	0.00
B4	0.39	1.54	0.09	0.00
A5	0.52	1.38	0.02	0.00
C1	0.26	1.87	0.04	0.00
C2	0.21	2.07	0.06	0.00
C3	0.25	1.78	0.05	0.00
C4	0.20	2.01	0.05	0.00
C5	0.20	2.02	0.06	0.00
Avg.	0.26	1.84	0.06	0.00

$$Loss = (Observed - Predicted)^2$$

Table 14 Rhodium recovery from fourteen samples of UG2 chromitite milled to 80% <75µm.

A2	<i>Rh content ppm</i>	<i>Rh content cumulative ppm</i>	<i>Rh distribution %</i>	<i>Rh distribution cumulative %</i>
RC1	10.78	10.78	71.89	71.89
RC2	2.03	6.13	15.38	87.28
RC3-5	1.04	3.80	12.50	99.77
RT	0.00	0.40	0.23	100.00

Rh calculated = 0.40 ppm Rh assay = 0.35ppm Difference = 0.05 ppm
Difference = 13%

B1	<i>Rh content ppm</i>	<i>Rh content cumulative ppm</i>	<i>Rh distribution %</i>	<i>Rh distribution cumulative %</i>
RC1	6.06	6.06	70.69	70.69
RC2	1.93	4.43	14.59	85.28
RC3-5	0.92	2.86	14.43	99.71
RT	0.00	0.31	0.29	100.00

Rh calculated = 0.31 ppm Rh assay = 0.37 ppm Difference = 0.06 ppm
Difference = 16%

A1	<i>Rh content ppm</i>	<i>Rh content cumulative ppm</i>	<i>Rh distribution %</i>	<i>Rh distribution cumulative %</i>
RC1	10.74	10.74	74.35	74.35
RC2	2.75	7.62	12.21	86.56
RC3-5	1.05	4.72	9.38	95.94
RT	0.02	0.54	4.06	100.00

Rh calculated = 0.54 ppm Rh assay = 0.43 ppm Difference = 0.11 ppm
Difference = 26%

B2	<i>Rh content ppm</i>	<i>Rh content cumulative ppm</i>	<i>Rh distribution %</i>	<i>Rh distribution cumulative %</i>
RC1	10.78	10.78	83.12	83.12
RC2	0.80	4.94	8.70	91.82
RC3-5	0.44	2.72	7.93	99.75
RT	0.00	0.35	0.25	100.00

Rh calculated = 0.35 ppm Rh assay = 0.33 ppm Difference = 0.02 ppm
Difference = 5%

A3	<i>Rh content ppm</i>	<i>Rh content cumulative ppm</i>	<i>Rh distribution %</i>	<i>Rh distribution cumulative %</i>
RC1	13.28	13.28	60.15	60.15
RC2	3.73	8.58	16.36	76.51
RC3-5	1.86	5.34	15.47	91.97
RT	0.06	0.66	8.03	100.00

Rh calculated = 0.66 ppm Rh assay = 0.60 ppm Difference = 0.06 ppm
Difference = 11%

Table 14 cont. Rhodium recovery from fourteen samples of UG2 chromitite milled to 80% <75µm.

B3	<i>Rh content ppm</i>	<i>Rh content cumulative ppm</i>	<i>Rh distribution %</i>	<i>Rh distribution cumulative %</i>
RC1	15.70	15.70	68.14	68.14
RC2	3.50	9.34	16.55	84.69
RC3-5	1.01	4.91	10.39	95.08
RT	0.03	0.55	4.92	100.00

*Rh calculated = 0.55 ppm Rh assay = 0.43 ppm Difference = 0.12 ppm
Difference = 21%*

A4	<i>Rh content ppm</i>	<i>Rh content cumulative ppm</i>	<i>Rh distribution %</i>	<i>Rh distribution cumulative %</i>
RC1	11.90	11.90	47.34	47.34
RC2	5.40	8.45	24.24	71.58
RC3-5	2.52	5.31	24.00	95.59
RT	0.03	0.57	4.41	100.00

*Rh calculated = 0.57 ppm Rh assay = 0.46 ppm Difference = 0.11 ppm
Difference = 20%*

B4	<i>Rh content ppm</i>	<i>Rh content cumulative ppm</i>	<i>Rh distribution %</i>	<i>Rh distribution cumulative %</i>
RC1	12.25	12.25	49.54	49.54
RC2	3.95	7.74	18.99	68.53
RC3-5	1.95	4.25	26.16	94.69
RT	0.04	0.57	5.31	100.00

*Rh calculated = 0.57 ppm Rh assay = 0.53 ppm Difference = 0.04 ppm
Difference = 8%*

A5	<i>Rh content ppm</i>	<i>Rh content cumulative ppm</i>	<i>Rh distribution %</i>	<i>Rh distribution cumulative %</i>
RC1	5.10	5.10	26.67	26.67
RC2	2.60	3.83	14.03	40.69
RC3-5	2.13	2.94	24.55	65.25
RT	0.22	0.56	34.75	100.00

*Rh calculated = 0.56 ppm Rh assay = 0.52 ppm Difference = 0.04 ppm
Difference = 6%*

C1	<i>Rh content ppm</i>	<i>Rh content cumulative ppm</i>	<i>Rh distribution %</i>	<i>Rh distribution cumulative %</i>
RC1	11.45	11.45	49.52	49.52
RC2	3.50	7.11	18.22	67.74
RC3-5	1.61	4.00	19.84	87.58
RT	0.08	0.53	12.42	100.00

*Rh calculated = 0.53 ppm Rh assay = 0.45 ppm Difference = 0.08 ppm
Difference = 17%*

Table 14 cont. Rhodium recovery from fourteen samples of UG2 chromitite milled to 80% <75 μ m.

C2	<i>Rh content ppm</i>	<i>Rh content cumulative ppm</i>	<i>Rh distribution %</i>	<i>Rh distribution cumulative %</i>
RC1	10.15	10.15	55.42	55.42
RC2	2.70	6.13	17.33	72.75
RC3-5	1.28	3.38	19.74	92.49
RT	0.05	0.51	7.51	100.00

Rh calculated = 0.51 ppm Rh assay = 0.47 ppm Difference = 0.04 ppm
Difference = 9%

C3	<i>Rh content ppm</i>	<i>Rh content cumulative ppm</i>	<i>Rh distribution %</i>	<i>Rh distribution cumulative %</i>
RC1	13.90	13.90	51.22	51.22
RC2	4.33	8.78	18.29	69.51
RC3-5	2.09	5.07	20.47	89.99
RT	0.08	0.66	10.01	100.00

Rh calculated = 0.66 ppm Rh assay = 0.55 ppm Difference = 0.11 ppm
Difference = 17%

C4	<i>Rh content ppm</i>	<i>Rh content cumulative ppm</i>	<i>Rh distribution %</i>	<i>Rh distribution cumulative %</i>
RC1	11.30	11.30	66.61	66.61
RC2	1.90	5.96	14.71	81.32
RC3-5	0.96	3.32	14.51	95.83
RT	0.03	0.51	4.17	100.00

Rh calculated = 0.51 ppm Rh assay = 0.47 ppm Difference = 0.04 ppm
Difference = 8%

C5	<i>Rh content ppm</i>	<i>Rh content cumulative ppm</i>	<i>Rh distribution %</i>	<i>Rh distribution cumulative %</i>
RC1	15.55	15.55	61.66	61.66
RC2	3.55	8.83	17.91	79.57
RC3-5	1.21	4.30	16.01	95.57
RT	0.03	0.50	4.43	100.00

Rh calculated = 0.50 ppm Rh assay = 0.50 ppm Difference = 0.00 ppm
Difference = 1%



Table 15 Kelsall flotation constants for rhodium recovery from fourteen samples of UG2 chromitite milled to 80% <75µm assuming an ultimate recovery of 100%.

Rh	ϕ	K_f	K_s	<i>Loss</i>
A2	0.30	2.13	0.25	0.00
B2	0.15	2.92	0.21	0.00
A1	0.21	2.24	0.25	0.00
B1	0.29	2.46	0.23	0.00
A3	0.29	1.63	0.06	0.00
B3	0.18	1.70	0.06	0.00
A4	0.36	1.16	0.11	0.00
B4	0.42	1.54	0.10	0.00
A5	0.64	1.15	0.03	0.00
C1	0.37	1.40	0.05	0.00
C2	0.33	1.59	0.07	0.00
C3	0.36	1.45	0.06	0.00
C4	0.24	1.89	0.09	0.00
C5	0.26	1.62	0.09	0.00
Avg.	0.31	1.78	0.12	0.00

$$Loss = (Observed - Predicted)^2$$

APPENDIX K

Multiple regression analysis results

- Table 1a** Observed and predicted values for R_f with % non-sulphide PGE mineral, PGE mineral grain diameter prior to milling, $pn/(pn+mil)$ ratio and the predicted PGE mineral degree of liberation as independent variables, in turn excluding one of the samples.
- Table 1b** Regression summary for dependent variable R_f with % non-sulphide PGE mineral, PGE mineral grain diameter prior to milling, $pn/(pn+mil)$ ratio and the predicted PGE mineral degree of liberation as independent variables, in turn excluding one of the samples.
- Table 2a** Observed and predicted values for R_f with % non-sulphide PGE mineral, $pn/(pn+mil)$ ratio and the predicted PGE mineral degree of liberation as independent variables, in turn excluding one of the samples.
- Table 2b** Regression summary for dependent variable R_f with % non-sulphide PGE mineral, $pn/(pn+mil)$ ratio, and the predicted PGE mineral degree of liberation as independent variables, in turn excluding one of the samples.
- Table 3a** Observed and predicted values for R_s with % non-sulphide PGE mineral, PGE mineral grain diameter prior to milling, $pn/(pn+mil)$ ratio, median chromite grain diameter and the predicted PGE mineral degree of liberation as independent variables, in turn excluding one of the samples.
- Table 3b** Regression summary for dependent variable R_s with % non-sulphide PGE mineral, PGE mineral grain diameter prior to milling, $pn/(pn+mil)$ ratio, median chromite grain diameter and the predicted PGE mineral degree of liberation as independent variables, in turn excluding one of the samples.
- Table 4a** Observed and predicted values for R_s with % non-sulphide PGE mineral, PGE mineral grain diameter prior to milling, $pn/(pn+mil)$ ratio, median chromite grain diameter and the predicted PGE mineral degree of liberation as independent variables, in turn excluding one of the samples.
- Table 4b** Regression summary for dependent variable R_s with % non-sulphide PGE mineral, $pn/(pn+mil)$ ratio and the predicted PGE mineral degree of

liberation as independent variables, in turn excluding sample A5 and one other sample.

Table 5 Regression summary for dependent variable 100-U with sample A5 excluded.

Table 6 Regression summary for dependent variable k_f . All fourteen samples included.

Table 7 Regression summary for dependent variable k_s . All fourteen samples included.

Table 8a Observed and predicted values for R_s with % non-sulphide PGE mineral and the PGE mineral degree of liberation as independent variables as independent variables, in turn excluding one of the samples.

Table 8b Regression summary for dependent variable R_f with % non-sulphide PGE mineral and the PGE mineral degree of liberation as independent variables, in turn excluding one sample.

Table 9a Observed and predicted values for R_f with % non-sulphide PGE mineral PGE mineral degree of liberation, base-metal sulphide degree of liberation, PGE mineral grain size prior, pentlandite/(pentlandite+millerite), and the PGE mineral degree of liberation as independent variables, in turn excluding one of the samples.

Table 9b Regression summary for dependent variable R_f with % non-sulphide PGE mineral, PGE mineral degree of liberation, base-metal sulphide degree of liberation, PGE mineral grain size prior, pentlandite/(pentlandite+millerite) to milling as independent variables, in turn excluding one sample.

Table 10a Observed and predicted values for variable R_s with % non-sulphide PGE mineral, PGE mineral degree of liberation, chromite grain diameter and PGE mineral diameter prior to milling, as independent variables, in turn excluding one of the samples.

Table 10b Regression summary for dependent variable R_s with % non-sulphide PGE mineral, PGE mineral degree of liberation, chromite grain diameter and PGE mineral diameter prior to milling as independent variables, in turn excluding one sample.

Table 11 Regression summary for dependent variable 100-U excluding sample A5.

Table 12 Regression summary for dependent variable k_f . All fourteen samples included.

Table 13 Regression summary for dependent variable k_s . All fourteen samples included.

Table 1a Observed and predicted values for R_f with % non-sulphide PGE mineral, PGE mineral grain diameter prior to milling, $pn/(pn+mil)$ ratio and the predicted PGE mineral degree of liberation as independent variables, in turn excluding one of the samples.

Sample number	R_f Observed	R_f Predicted (n=14)	* R_f Predicted (n=13)	R_f Observed – * R_f predicted
A2	75.1	75.1	75.1	0.0
B2	80.3	79.3	79.0	1.4
A1	79.8	79.1	78.8	1.0
B1	73.8	76.3	77.7	-3.8
A3	63.4	65.3	67.3	-3.9
B3	74.4	71.9	70.9	3.5
A4	59.9	61.2	62.6	-2.7
B4	56.5	55.3	53.0	3.5
A5	39.7	39.3	37.2	2.5
C1	66.3	69.3	70.1	-3.8
C2	71.5	69.8	69.4	2.1
C3	68.2	68.8	68.4	-0.2
C4	74.8	72.3	71.6	3.2
C5	74.8	75.6	75.8	-1.0

* R_f predicted : Value of R_f determined from the regression equation calculated from the other 13 samples.

Table 1b Regression summary for dependent variable R_f with % non-sulphide PGE mineral, PGE mineral grain diameter prior to milling, $pn/(pn+mil)$ ratio and the predicted PGE mineral degree of liberation as independent variables, in turn excluding one of the samples.

Regression Summary for Dependent Variable: R_f A2 excluded				
R= .98881140 R ² = .97774799 Adjusted R ² = .96662199				
F(4,8)=87.880 p<.00000 Std.Error of estimate: 2.0388				
	B	St. Err. of B	t(8)	p-level
Intercept	61.67	7.23	8.53	0.00
% non sulphide PGEM	-0.22	0.03	-7.04	0.00
PGEM grain diameter at <2mm	-7.18	2.15	-3.34	0.01
$pn/(pn+mil)$	-22.32	3.20	-6.97	0.00
PGEM pred. degree of liberation	0.68	0.07	10.03	0.00
Regression Summary for Dependent Variable: R_f B2 excluded				
R= .98827751 R ² = .97669244 Adjusted R ² = .96503866				
F(4,8)=83.809 p<.00000 Std.Error of estimate: 2.0123				
	B	St. Err. of B	t(8)	p-level
Intercept	61.16	7.08	8.63	0.00
% non sulphide PGEM	-0.21	0.03	-6.93	0.00
PGEM grain diameter at <2mm	-6.95	2.12	-3.27	0.01
$pn/(pn+mil)$	-22.25	3.03	-7.35	0.00
PGEM pred. degree of liberation	0.68	0.07	10.01	0.00

Table 1b continued Regression summary for dependent variable R_f with % non-sulphide PGE mineral, PGE mineral grain diameter prior to milling, $pn/(pn+mil)$ ratio, and the predicted PGE mineral degree of liberation as independent variables, in turn excluding one of the samples.

Regression Summary for Dependent Variable: R_f A1 excluded				
R= .98811817 R ² = .97637751 Adjusted R ² = .96456626 F(4,8)=82.665 p<.00000 Std.Error of estimate: 2.0356				
	B	St. Err. of B	t(8)	p-level
Intercept	61.94	7.33	8.45	0.00
% non sulphide PGEM	-0.22	0.03	-7.18	0.00
PGEM grain diameter at <2mm	-7.23	2.11	-3.42	0.01
$pn/(pn+mil)$	-22.30	3.06	-7.29	0.00
PGEM pred. degree of liberation	0.68	0.07	9.63	0.00
Regression Summary for Dependent Variable: R_f B1 excluded				
R= .99190668 R ² = .98387886 Adjusted R ² = .97581830 F(4,8)=122.06 p<.00000 Std.Error of estimate: 1.7447				
	B	St. Err. of B	t(8)	p-level
Intercept	65.46	6.47	10.12	0.00
% non sulphide PGEM	-0.22	0.03	-8.55	0.00
PGEM grain diameter at <2mm	-8.85	2.04	-4.34	0.00
$pn/(pn+mil)$	-23.42	2.70	-8.66	0.00
PGEM pred. degree of liberation	0.70	0.06	11.81	0.00
Regression Summary for Dependent Variable: R_f A3 excluded				
R= .99126096 R ² = .98259828 Adjusted R ² = .97389742 F(4,8)=112.93 p<.00000 Std.Error of estimate: 1.8146				
	B	St. Err. of B	t(8)	p-level
Intercept	58.94	6.59	8.95	0.00
% non sulphide PGEM	-0.23	0.03	-8.14	0.00
PGEM grain diameter at <2mm	-5.56	2.17	-2.57	0.03
$pn/(pn+mil)$	-19.04	3.54	-5.38	0.00
PGEM pred. degree of liberation	0.64	0.07	9.50	0.00
Regression Summary for Dependent Variable: R_f B3 excluded				
R= .99284297 R ² = .98573717 Adjusted R ² = .97860575 F(4,8)=138.22 p<.00000 Std.Error of estimate: 1.6374				
	B	St. Err. of B	t(8)	p-level
Intercept	56.61	6.19	9.15	0.00
% non sulphide PGEM	-0.20	0.03	-7.59	0.00
PGEM grain diameter at <2mm	-5.72	1.82	-3.15	0.01
$pn/(pn+mil)$	-22.96	2.48	-9.25	0.00
PGEM pred. degree of liberation	0.70	0.06	12.66	0.00

Table 1b continued Regression summary for dependent variable R_f with % non-sulphide PGE mineral, PGE mineral grain diameter prior to milling, $pn/(pn+mil)$ ratio and the predicted PGE mineral degree of liberation as independent variables, in turn excluding one of the samples.

Regression Summary for Dependent Variable: R_f A4 excluded				
R= .98942983 R ² = .97897140 Adjusted R ² = .96845709 F(4,8)=93.109 p<.00000 Std.Error of estimate: 1.9610				
	B	St. Err. of B	t(8)	p-level
Intercept	60.96	6.88	8.86	0.00
% non sulphide PGEM	-0.20	0.04	-5.49	0.00
PGEM grain diameter at <2mm	-7.46	2.04	-3.65	0.01
$pn/(pn+mil)$	-23.73	3.44	-6.90	0.00
PGEM pred. degree of liberation	0.71	0.08	9.41	0.00
Regression Summary for Dependent Variable: R_f B4 excluded				
R= .99023976 R ² = .98057479 Adjusted R ² = .97086218 F(4,8)=100.96 p<.00000 Std.Error of estimate: 1.8356				
	B	St. Err. of B	t(8)	p-level
Intercept	70.81	9.26	7.65	0.00
% non sulphide PGEM	-0.26	0.04	-6.20	0.00
PGEM grain diameter at <2mm	-9.64	2.61	-3.70	0.01
$pn/(pn+mil)$	-21.92	2.77	-7.90	0.00
PGEM pred. degree of liberation	0.64	0.07	9.50	0.00
Regression Summary for Dependent Variable: R_f A5 excluded				
R= .97404853 R ² = .94877054 Adjusted R ² = .92315580 F(4,8)=37.040 p<.00003 Std.Error of estimate: 2.0382				
	B	St. Err. of B	t(8)	p-level
Intercept	61.07	11.24	5.44	0.00
% non sulphide PGEM	-0.22	0.03	-6.81	0.00
PGEM grain diameter at <2mm	-7.10	2.35	-3.02	0.02
$pn/(pn+mil)$	-22.55	4.84	-4.66	0.00
PGEM pred. degree of liberation	0.69	0.14	4.96	0.00
Regression Summary for Dependent Variable: R_f C1 excluded				
R= .99065733 R ² = .98140195 Adjusted R ² = .97210293 F(4,8)=105.54 p<.00000 Std.Error of estimate: 1.8900				
	B	St. Err. of B	t(8)	p-level
Intercept	61.61	6.58	9.37	0.00
% non sulphide PGEM	-0.21	0.03	-7.62	0.00
PGEM grain diameter at <2mm	-6.93	1.95	-3.56	0.01
$pn/(pn+mil)$	-22.71	2.86	-7.93	0.00
PGEM pred. degree of liberation	0.68	0.06	10.77	0.00

Table 1b continued Regression summary for dependent variable R_f with % non-sulphide PGE mineral, PGE mineral grain diameter prior to milling, $pn/(pn+mil)$ ratio and the predicted PGE mineral degree of liberation as independent variables, in turn excluding one of the samples.

Regression Summary for Dependent Variable: R_f C2 excluded				
R= .98953456 R ² = .97917865 Adjusted R ² = .96876798 F(4,8)=94.055 p<.00000 Std.Error of estimate: 1.9967				
	B	St. Err. of B	t(8)	p-level
Intercept	61.25	6.98	8.77	0.00
% non sulphide PGEM	-0.22	0.03	-7.30	0.00
PGEM grain diameter at <2mm	-7.17	2.05	-3.51	0.01
$pn/(pn+mil)$	-22.07	3.03	-7.29	0.00
PGEM pred. degree of liberation	0.68	0.07	10.26	0.00
Regression Summary for Dependent Variable: R_f C3 excluded				
R= .98937372 R ² = .97886036 Adjusted R ² = .96829054 F(4,8)=92.609 p<.00000 Std.Error of estimate: 2.0182				
	B	St. Err. of B	t(8)	p-level
Intercept	61.87	7.05	8.78	0.00
% non sulphide PGEM	-0.22	0.03	-7.25	0.00
PGEM grain diameter at <2mm	-7.14	2.07	-3.45	0.01
$pn/(pn+mil)$	-22.33	3.04	-7.36	0.00
PGEM pred. degree of liberation	0.68	0.07	10.02	0.00
Regression Summary for Dependent Variable: R_f C4 excluded				
R= .99114362 R ² = .98236568 Adjusted R ² = .97354851 F(4,8)=111.42 p<.00000 Std.Error of estimate: 1.8170				
	B	St. Err. of B	t(8)	p-level
Intercept	61.31	6.33	9.69	0.00
% non sulphide PGEM	-0.22	0.03	-8.09	0.00
PGEM grain diameter at <2mm	-7.21	1.86	-3.87	0.00
$pn/(pn+mil)$	-21.43	2.80	-7.66	0.00
PGEM pred. degree of liberation	0.68	0.06	11.15	0.00
Regression Summary for Dependent Variable: R_f C5 excluded				
R= .98922510 R ² = .97856629 Adjusted R ² = .96784943 F(4,8)=91.311 p<.00000 Std.Error of estimate: 2.0037				
	B	St. Err. of B	t(8)	p-level
Intercept	62.18	7.05	8.83	0.00
% non sulphide PGEM	-0.22	0.03	-7.30	0.00
PGEM grain diameter at <2mm	-7.35	2.08	-3.53	0.01
$pn/(pn+mil)$	-22.80	3.16	-7.23	0.00
PGEM pred. degree of liberation	0.69	0.07	10.18	0.00

Table 2a Observed and predicted values for R_f with % non-sulphide PGE mineral, $pn/(pn+mil)$ ratio and the predicted PGE mineral degree of liberation as independent variables, in turn excluding one of the samples.

Sample number	R_f Observed	R_f Predicted (n=14)	* R_f Predicted (n=13)	R_f Observed – * R_f predicted
A2	75.1	76.5	76.9	-1.8
B2	80.3	77.7	77.0	3.3
A1	79.8	80.2	80.3	-0.5
B1	73.8	73.0	72.9	0.9
A3	63.4	68.3	70.9	-7.5
B3	74.4	69.1	68.2	6.2
A4	59.9	60.5	61.1	-1.2
B4	56.5	59.6	41.7	14.8
A5	39.7	37.8	30.4	9.3
C1	66.3	69.7	70.7	-4.4
C2	71.5	70.0	69.6	1.9
C3	68.2	69.1	69.3	-1.1
C4	74.8	72.5	71.8	3.1
C5	74.8	74.6	68.5	6.3

* R_f predicted : Value of R_f determined from the regression equation calculated from the other 13 samples.

Table 2b Regression summary for dependent variable R_f with % non-sulphide PGE mineral, $pn/(pn+mil)$ ratio, and the predicted PGE mineral degree of liberation as independent variables, in turn excluding one of the samples.

Regression Summary for Dependent Variable: R_f A2 excluded				
R= .97297118 R ² = .94667292 Adjusted R ² = .92889723				
F(3,9)=53.257 p<.00000 Std.Error of estimate: 2.9756				
	B	St. Err. of B	t(9)	p-level
Intercept	40.24	4.87	8.26	0.00
% non sulphide PGEM	-0.18	0.04	-4.26	0.00
$pn/(pn+mil)$	-18.63	4.38	-4.25	0.00
PGEM pred. degree of liberation	0.69	0.10	6.98	0.00
Regression Summary for Dependent Variable: R_f B2 excluded				
R= .97234955 R ² = .94546365 Adjusted R ² = .92728486				
F(3,9)=52.009 p<.00001 Std.Error of estimate: 2.9020				
	B	St. Err. of B	t(9)	p-level
Intercept	40.67	4.78	8.50	0.00
% non sulphide PGEM	-0.16	0.04	-4.24	0.00
$pn/(pn+mil)$	-19.28	4.17	-4.63	0.00
PGEM pred. degree of liberation	0.68	0.10	6.99	0.00

Table 2b continued Regression summary for dependent variable R_f with % non-sulphide PGE mineral, $pn/(pn+mil)$ ratio, and the predicted PGE mineral degree of liberation as independent variables, in turn excluding one of the samples.

Regression Summary for Dependent Variable: R_f A1 excluded				
R= .97048122 R ² = .94183380 Adjusted R ² = .92244507 F(3,9)=48.576 p<.00001 Std.Error of estimate: 3.0116				
	B	St. Err. of B	t(9)	p-level
Intercept	39.81	5.09	7.82	0.00
% non sulphide PGEM	-0.17	0.04	-4.27	0.00
$pn/(pn+mil)$	-19.22	4.33	-4.44	0.00
PGEM pred. degree of liberation	0.70	0.10	6.75	0.00
Regression Summary for Dependent Variable: R_f B1 excluded				
R= .97258564 R ² = .94592283 Adjusted R ² = .92789710 F(3,9)=52.476 p<.00001 Std.Error of estimate: 3.0127				
	B	St. Err. of B	t(9)	p-level
Intercept	40.41	5.04	8.01	0.00
% non sulphide PGEM	-0.17	0.04	-4.27	0.00
$pn/(pn+mil)$	-19.08	4.34	-4.40	0.00
PGEM pred. degree of liberation	0.69	0.10	6.70	0.00
Regression Summary for Dependent Variable: R_f A3 excluded				
R= .98400607 R ² = .96826795 Adjusted R ² = .95769060 F(3,9)=91.542 p<.00000 Std.Error of estimate: 2.3103				
	B	St. Err. of B	t(9)	p-level
Intercept	44.19	4.10	10.77	0.00
% non sulphide PGEM	-0.21	0.03	-6.09	0.00
$pn/(pn+mil)$	-14.28	3.84	-3.72	0.00
PGEM pred. degree of liberation	0.61	0.08	7.22	0.00
Regression Summary for Dependent Variable: R_f B3 excluded				
R= .97337194 R ² = .94745294 Adjusted R ² = .92993725 F(3,9)=54.092 p<.00000 Std.Error of estimate: 3.0000				
	B	St. Err. of B	t(9)	p-level
Intercept	40.59	5.05	8.03	0.00
% non sulphide PGEM	-0.17	0.04	-4.30	0.00
$pn/(pn+mil)$	-19.23	4.31	-4.46	0.00
PGEM pred. degree of liberation	0.69	0.10	6.86	0.00
Regression Summary for Dependent Variable: R_f A4 excluded				
R= .97571419 R ² = .95201819 Adjusted R ² = .93602425 F(3,9)=59.524 p<.00000 Std.Error of estimate: 2.8621				
	B	St. Err. of B	t(9)	p-level
Intercept	41.08	4.77	8.61	0.00
% non sulphide PGEM	-0.17	0.04	-4.45	0.00
$pn/(pn+mil)$	-19.85	4.16	-4.77	0.00
PGEM pred. degree of liberation	0.69	0.10	7.23	0.00

Table 2b continued Regression summary for dependent variable R_f with % non-sulphide PGE mineral, $pn/(pn+mil)$ ratio, and the predicted PGE mineral degree of liberation as independent variables, in turn excluding one of the samples.

Regression Summary for Dependent Variable: R_f B4 excluded				
R= .97435099 R ² = .94935986 Adjusted R ² = .93247981 F(3,9)=56.242 p<.00000 Std.Error of estimate: 2.9029				
	B	St. Err. of B	t(9)	p-level
Intercept	39.72	4.77	8.32	0.00
% non sulphide PGEM	-0.17	0.04	-4.45	0.00
$pn/(pn+mil)$	-18.31	4.28	-4.28	0.00
PGEM pred. degree of liberation	0.69	0.10	7.12	0.00
Regression Summary for Dependent Variable: R_f A5 excluded				
R= .97323841 R ² = .94719300 Adjusted R ² = .92959067 F(3,9)=53.811 p<.00000 Std.Error of estimate: 2.9980				
	B	St. Err. of B	t(9)	p-level
Intercept	39.75	5.01	7.94	0.00
% non sulphide PGEM	-0.17	0.04	-4.27	0.00
$pn/(pn+mil)$	-18.94	4.34	-4.36	0.00
PGEM pred. degree of liberation	0.70	0.10	6.96	0.00
Regression Summary for Dependent Variable: R_f C1 excluded				
R= .97217230 R ² = .94511898 Adjusted R ² = .92682530 F(3,9)=51.664 p<.00001 Std.Error of estimate: 3.0229				
	B	St. Err. of B	t(9)	p-level
Intercept	40.14	4.94	8.12	0.00
% non sulphide PGEM	-0.17	0.04	-4.24	0.00
$pn/(pn+mil)$	-19.14	4.50	-4.26	0.00
PGEM pred. degree of liberation	0.69	0.10	6.82	0.00
Regression Summary for Dependent Variable: R_f C2 excluded				
R= .97152886 R ² = .94386833 Adjusted R ² = .92515777 F(3,9)=50.446 p<.00001 Std.Error of estimate: 3.0206				
	B	St. Err. of B	t(9)	p-level
Intercept	39.80	5.72	6.96	0.00
% non sulphide PGEM	-0.17	0.05	-3.06	0.01
$pn/(pn+mil)$	-19.46	4.98	-3.90	0.00
PGEM pred. degree of liberation	0.70	0.12	6.01	0.00
Regression Summary for Dependent Variable: R_f C3 excluded				
R= .97337194 R ² = .94745294 Adjusted R ² = .92993725 F(3,9)=54.092 p<.00000 Std.Error of estimate: 3.0000				
	B	St. Err. of B	t(9)	p-level
Intercept	40.59	5.05	8.03	0.00
% non sulphide PGEM	-0.17	0.04	-4.30	0.00
$pn/(pn+mil)$	-19.23	4.31	-4.46	0.00
PGEM pred. degree of liberation	0.69	0.10	6.86	0.00

Table 2b continued Regression summary for dependent variable R_f with % non-sulphide PGE mineral, $pn/(pn+mil)$ ratio, and the predicted PGE mineral degree of liberation as independent variables, in turn excluding one of the samples.

Regression Summary for Dependent Variable: R_f C4 excluded				
R= .98387705 R ² = .96801404 Adjusted R ² = .95735206 F(3,9)=90.791 p<.00000 Std.Error of estimate: 2.3118				
	B	St. Err. of B	t(9)	p-level
Intercept	39.06	3.81	10.26	0.00
% non sulphide PGEM	-0.15	0.03	-4.95	0.00
$pn/(pn+mil)$	-21.20	3.41	-6.21	0.00
PGEM pred. degree of liberation	0.72	0.08	9.26	0.00
Regression Summary for Dependent Variable: R_f C5 excluded				
R= .97333239 R ² = .94737593 Adjusted R ² = .92983458 F(3,9)=54.008 p<.00000 Std.Error of estimate: 2.8485				
	B	St. Err. of B	t(9)	p-level
Intercept	38.61	4.88	7.92	0.00
% non sulphide PGEM	-0.15	0.04	-3.32	0.01
$pn/(pn+mil)$	-20.44	4.26	-4.80	0.00
PGEM pred. degree of liberation	0.72	0.10	7.31	0.00

Table 3a Observed and predicted values for R_s with % non-sulphide PGE mineral, PGE mineral grain diameter prior to milling, $pn/(pn+mil)$ ratio, median chromite grain diameter and the predicted PGE mineral degree of liberation as independent variables, in turn excluding one of the samples.

Sample number	R_s Observed	R_s Predicted (n=14)	* R_s Predicted (n=13)	R_s Observed – * R_s predicted
A2	18.8	18.0	17.7	1.1
B2	15.3	15.9	16.1	-0.8
A1	16.3	17.7	18.5	-2.2
B1	19.6	18.7	18.3	1.4
A3	25.9	25.5	24.9	1.0
B3	21.0	21.1	21.1	-0.1
A4	33.7	33.7	33.7	0.0
B4	37.7	37.5	37.2	0.5
A5	28.7	29.0	37.1	-8.4
C1	24.5	24.9	25.1	-0.6
C2	22.6	23.4	19.5	3.0
C3	24.0	23.8	23.7	0.3
C4	19.9	18.8	17.8	2.1
C5	21.1	21.1	21.1	0.0

* R_s predicted : Value of R_s determined from the regression equation calculated for the other 13 samples.

Table 3b Regression summary for dependent variable R_s with % non-sulphide PGE mineral, PGE mineral grain diameter prior to milling, $pn/(pn+mil)$ ratio, median chromite grain diameter and the predicted PGE mineral degree of liberation as independent variables, in turn excluding one of the samples.

Regression Summary for Dependent Variable: R_s A2 excluded				
R= .99492020 R ² = .98986621 Adjusted R ² = .98262778 F(5,7)=136.75 p<.00000 Std.Error of estimate: .84867				
	B	St. Err. of B	t(7)	p-level
Intercept	-5.26	3.02	-1.74	0.13
% non sulphide PGEM	0.17	0.02	9.90	0.00
PGEM grain diameter at <2mm	5.85	1.03	5.68	0.00
Chromite grain diameter	0.11	0.01	8.02	0.00
$pn/(pn+mil)$	4.86	1.45	3.35	0.01
PGEM pred. degree of liberation	-0.20	0.04	-5.34	0.00
Regression Summary for Dependent Variable: R_s B2 excluded				
R= .99425144 R ² = .98853592 Adjusted R ² = .98034730 F(5,7)=120.72 p<.00000 Std.Error of estimate: .85769				
	B	St. Err. of B	t(7)	p-level
Intercept	-5.46	3.02	-1.81	0.11
% non sulphide PGEM	0.17	0.02	9.77	0.00
PGEM grain diameter at <2mm	6.03	0.99	6.07	0.00
Chromite grain diameter	0.11	0.01	7.74	0.00
$pn/(pn+mil)$	4.91	1.46	3.36	0.01
PGEM pred. degree of liberation	-0.19	0.04	-4.94	0.00
Regression Summary for Dependent Variable: R_s A1 excluded				
R= .99669983 R ² = .99341055 Adjusted R ² = .98870380 F(5,7)=211.06 p<.00000 Std.Error of estimate: .66260				
	B	St. Err. of B	t(7)	p-level
Intercept	-7.44	2.42	-3.08	0.02
% non sulphide PGEM	0.18	0.01	12.87	0.00
PGEM grain diameter at <2mm	6.76	0.81	8.36	0.00
Chromite grain diameter	0.10	0.01	8.48	0.00
$pn/(pn+mil)$	4.49	1.14	3.92	0.01
PGEM pred. degree of liberation	-0.16	0.03	-4.88	0.00
Regression Summary for Dependent Variable: R_s B1 excluded				
R= .99556347 R ² = .99114663 Adjusted R ² = .98482279 F(5,7)=156.73 p<.00000 Std.Error of estimate: .79951				
	B	St. Err. of B	t(7)	p-level
Intercept	-7.06	2.97	-2.38	0.05
% non sulphide PGEM	0.17	0.02	10.62	0.00
PGEM grain diameter at <2mm	6.70	1.03	6.53	0.00
Chromite grain diameter	0.11	0.01	8.56	0.00
$pn/(pn+mil)$	5.40	1.38	3.91	0.01
PGEM pred. degree of liberation	-0.20	0.04	-5.73	0.00

Table 3b continued Regression summary for dependent variable R_s with % non-sulphide PGE mineral, PGE mineral grain diameter prior to milling, $pn/(pn+mil)$ ratio, median chromite grain diameter and the predicted PGE mineral degree of liberation as independent variables, in turn excluding one of the samples.

Regression Summary for Dependent Variable: R_s A3 excluded				
R= .99542266 R ² = .99086628 Adjusted R ² = .98434219 F(5,7)=151.88 p<.00000 Std.Error of estimate: .82015				
	B	St. Err. of B	t(7)	p-level
Intercept	-4.77	3.00	-1.59	0.16
% non sulphide PGEM	0.18	0.02	9.14	0.00
PGEM grain diameter at <2mm	5.76	1.00	5.78	0.00
Chromite grain diameter	0.10	0.02	6.56	0.00
$pn/(pn+mil)$	3.32	2.11	1.57	0.16
PGEM pred. degree of liberation	-0.16	0.05	-3.50	0.01
Regression Summary for Dependent Variable: R_s B3 excluded				
R= .99470937 R ² = .98944674 Adjusted R ² = .98190870 F(5,7)=131.26 p<.00000 Std.Error of estimate: .88096				
	B	St. Err. of B	t(7)	p-level
Intercept	-5.31	3.46	-1.53	0.17
% non sulphide PGEM	0.17	0.02	7.54	0.00
PGEM grain diameter at <2mm	5.91	1.24	4.79	0.00
Chromite grain diameter	0.11	0.02	6.83	0.00
$pn/(pn+mil)$	5.21	1.61	3.24	0.01
PGEM pred. degree of liberation	-0.20	0.04	-4.62	0.00
Regression Summary for Dependent Variable: R_s A4 excluded				
R= .99329542 R ² = .98663580 Adjusted R ² = .97708994 F(5,7)=103.36 p<.00000 Std.Error of estimate: .88413				
	B	St. Err. of B	t(7)	p-level
Intercept	-5.70	3.10	-1.84	0.11
% non sulphide PGEM	0.17	0.02	8.88	0.00
PGEM grain diameter at <2mm	6.17	1.06	5.83	0.00
Chromite grain diameter	0.11	0.01	7.40	0.00
$pn/(pn+mil)$	5.15	1.62	3.17	0.02
PGEM pred. degree of liberation	-0.20	0.04	-4.94	0.00
Regression Summary for Dependent Variable: R_s B4 excluded				
R= .99102283 R ² = .98212625 Adjusted R ² = .96935928 F(5,7)=76.927 p<.00001 Std.Error of estimate: .88276				
	B	St. Err. of B	t(7)	p-level
Intercept	-5.01	4.45	-1.13	0.30
% non sulphide PGEM	0.17	0.02	7.17	0.00
PGEM grain diameter at <2mm	5.92	1.32	4.47	0.00
Chromite grain diameter	0.11	0.01	7.77	0.00
$pn/(pn+mil)$	5.06	1.50	3.38	0.01
PGEM pred. degree of liberation	-0.20	0.04	-4.89	0.00

Table 3b continued Regression summary for dependent variable R_s with % non-sulphide PGE mineral, PGE mineral grain diameter prior to milling, $pn/(pn+mil)$ ratio, median chromite grain diameter and the predicted PGE mineral degree of liberation as independent variables, in turn excluding one of the samples.

Regression Summary for Dependent Variable: R_s A5 excluded				
R= .99684844 R ² = .99370681 Adjusted R ² = .98921168 F(5,7)=221.06 p<.00000 Std.Error of estimate: .66522				
	B	St. Err. of B	t(7)	p-level
Intercept	7.10	5.99	1.19	0.27
% non sulphide PGEM	0.19	0.02	11.67	0.00
PGEM grain diameter at <2mm	5.86	0.77	7.59	0.00
Chromite grain diameter	0.07	0.02	3.13	0.02
$pn/(pn+mil)$	8.66	1.92	4.51	0.00
PGEM pred. degree of liberation	-0.30	0.05	-5.60	0.00
Regression Summary for Dependent Variable: R_s C1 excluded				
R= .99615900 R ² = .99233275 Adjusted R ² = .98685614 F(5,7)=181.19 p<.00000 Std.Error of estimate: .75508				
	B	St. Err. of B	t(7)	p-level
Intercept	-5.74	2.63	-2.18	0.07
% non sulphide PGEM	0.17	0.02	10.96	0.00
PGEM grain diameter at <2mm	6.12	0.87	7.05	0.00
Chromite grain diameter	0.12	0.01	9.23	0.00
$pn/(pn+mil)$	5.01	1.28	3.93	0.01
PGEM pred. degree of liberation	-0.21	0.03	-6.13	0.00
Regression Summary for Dependent Variable: R_s C2 excluded				
R= .99488254 R ² = .98979127 Adjusted R ² = .98249931 F(5,7)=135.74 p<.00000 Std.Error of estimate: .87141				
	B	St. Err. of B	t(7)	p-level
Intercept	-5.61	3.05	-1.84	0.11
% non sulphide PGEM	0.17	0.02	9.58	0.00
PGEM grain diameter at <2mm	6.09	1.00	6.08	0.00
Chromite grain diameter	0.11	0.01	7.89	0.00
$pn/(pn+mil)$	4.99	1.48	3.38	0.01
PGEM pred. degree of liberation	-0.20	0.04	-5.16	0.00
Regression Summary for Dependent Variable: R_s C3 excluded				
R= .99487254 R ² = .98977136 Adjusted R ² = .98246520 F(5,7)=135.47 p<.00000 Std.Error of estimate: .87287				
	B	St. Err. of B	t(7)	p-level
Intercept	-5.89	3.05	-1.93	0.10
% non sulphide PGEM	0.17	0.02	9.61	0.00
PGEM grain diameter at <2mm	6.14	1.00	6.12	0.00
Chromite grain diameter	0.11	0.01	7.68	0.00
$pn/(pn+mil)$	5.00	1.48	3.38	0.01
PGEM pred. degree of liberation	-0.19	0.04	-4.94	0.00

Table 3b continued Regression summary for dependent variable R_s with % non-sulphide PGE mineral, PGE mineral grain diameter prior to milling, $pn/(pn+mil)$ ratio, median chromite grain diameter and the predicted PGE mineral degree of liberation as independent variables, in turn excluding one of the samples.

Regression Summary for Dependent Variable: R_s C4 excluded				
R= .99630132 R ² =.99261631 Adjusted R ² = .98734225				
F(5,7)=188.21 p<.00000 Std.Error of estimate: .73171				
	B	St. Err. of B	t(7)	p-level
Intercept	-5.83	2.55	-2.29	0.06
% non sulphide PGEM	0.16	0.02	9.60	0.00
PGEM grain diameter at <2mm	5.65	0.88	6.42	0.00
Chromite grain diameter	0.13	0.01	8.83	0.00
$pn/(pn+mil)$	6.24	1.41	4.44	0.00
PGEM pred. degree of liberation	-0.23	0.04	-6.30	0.00
Regression Summary for Dependent Variable: R_s C5 excluded				
R= .99469230 R ² = .98941278 Adjusted R ² = .98185048				
F(5,7)=130.83 p<.00000 Std.Error of estimate: .88288				
	B	St. Err. of B	t(7)	p-level
Intercept	-5.87	3.11	-1.89	0.10
% non sulphide PGEM	0.17	0.02	9.54	0.00
PGEM grain diameter at <2mm	6.16	1.03	5.98	0.00
Chromite grain diameter	0.11	0.01	7.73	0.00
$pn/(pn+mil)$	5.12	1.53	3.34	0.01
PGEM pred. degree of liberation	-0.20	0.04	-5.08	0.00

Table 4a Observed and predicted values for R_s with % non-sulphide PGE mineral, PGE mineral grain diameter prior to milling, $pn/(pn+mil)$ ratio, median chromite grain diameter and the predicted PGE mineral degree of liberation as independent variables, in turn excluding one of the samples.

Sample number	R_s Observed	R_s Predicted (n=14)	* R_s Predicted (n=13)	R_s Observed – * R_s predicted
A2	18.8	17.3	16.8	2.0
B2	19.6	21.5	21.9	-2.3
A1	16.3	14.5	13.2	3.1
B1	15.3	16.6	17.0	-1.7
A3	25.9	24.4	24.0	1.9
B3	21.0	24.4	25.4	-4.4
A4	33.7	34.7	35.9	-2.2
B4	37.7	35.1	33.1	4.6
C1	24.5	23.6	23.3	1.3
C2	22.6	23.5	23.7	-1.1
C3	24.0	24.1	24.2	-0.2
C4	19.9	21.3	21.7	-1.8
C5	21.1	19.4	18.8	2.3

* R_s predicted : Value of R_s determined from the regression equation calculated for the other 13 samples.

Table 4b Regression summary for dependent variable R_s with % non-sulphide PGE mineral, $pn/(pn+mil)$ ratio and the predicted PGE mineral degree of liberation as independent variables, in turn excluding sample A5 and one other sample.

Regression Summary for Dependent Variable: R_s A2 excluded				
R= .96436815 R ² = .93000593 Adjusted R ² = .90375816 F(3,8)=35.432 p<.00006 Std.Error of estimate: 2.0319				
	B	St. Err. of B	t(8)	p-level
Intercept	46.80	5.87	7.98	0.00
% non sulphide PGEM	0.20	0.03	6.95	0.00
$pn/(pn+mil)$	14.04	4.76	2.95	0.02
PGEM pred. degree of liberation	-0.55	0.13	-4.41	0.00
Regression Summary for Dependent Variable: R_s B2 excluded				
R= .95989945 R ² = .92140695 Adjusted R ² = .89193455 F(3,8)=31.263 p<.00009 Std.Error of estimate: 2.0464				
	B	St. Err. of B	t(8)	p-level
Intercept	45.58	6.17	7.39	0.00
% non sulphide PGEM	0.19	0.03	6.86	0.00
$pn/(pn+mil)$	13.82	4.86	2.85	0.02
PGEM pred. degree of liberation	-0.53	0.13	-3.98	0.00
Regression Summary for Dependent Variable: R_s A1 excluded				
R= .96571593 R ² = .93260726 Adjusted R ² = .90733498 F(3,8)=36.902 p<.00005 Std.Error of estimate: 1.9311				
	B	St. Err. of B	t(8)	p-level
Intercept	50.79	6.49	7.82	0.00
% non sulphide PGEM	0.19	0.03	7.50	0.00
$pn/(pn+mil)$	16.89	4.93	3.43	0.01
PGEM pred. degree of liberation	-0.64	0.14	-4.59	0.00
Regression Summary for Dependent Variable: R_s B1 excluded				
R= .96697151 R ² = .93503391 Adjusted R ² = .91067162 F(3,8)=38.380 p<.00004 Std.Error of estimate: 1.9725				
	B	St. Err. of B	t(8)	p-level
Intercept	45.29	5.86	7.73	0.00
% non sulphide PGEM	0.19	0.03	7.34	0.00
$pn/(pn+mil)$	13.66	4.66	2.93	0.02
PGEM pred. degree of liberation	-0.52	0.13	-4.16	0.00
Regression Summary for Dependent Variable: R_s A3 excluded				
R= .96985665 R ² = .94062193 Adjusted R ² = .91835515 F(3,8)=42.243 p<.00003 Std.Error of estimate: 1.8949				
	B	St. Err. of B	t(8)	p-level
Intercept	39.31	7.74	5.08	0.00
% non sulphide PGEM	0.22	0.03	6.81	0.00
$pn/(pn+mil)$	7.69	6.64	1.16	0.28
PGEM pred. degree of liberation	-0.39	0.17	-2.34	0.05

Table 4b continued Regression summary for dependent variable R_s with % non-sulphide PGE mineral, $pn/(pn+mil)$ ratio and the predicted PGE mineral degree of liberation as independent variables, in turn excluding sample A5 and one other sample.

Regression Summary for Dependent Variable: R_s B3 excluded				
R= .97878183 R ² = .95801388 Adjusted R ² = .94226909 F(3,8)=60.846 p<.00001 Std.Error of estimate: 1.5993				
	B	St. Err. of B	t(8)	p-level
Intercept	50.45	4.87	10.37	0.00
% non sulphide PGEM	0.18	0.02	8.18	0.00
$pn/(pn+mil)$	18.12	4.03	4.50	0.00
PGEM pred. degree of liberation	-0.63	0.10	-6.07	0.00
Regression Summary for Dependent Variable: R_s A4 excluded				
R= .95388385 R ² = .90989441 Adjusted R ² = .87610481 F(3,8)=26.928 p<.00016 Std.Error of estimate: 2.0422				
	B	St. Err. of B	t(8)	p-level
Intercept	45.98	6.00	7.67	0.00
% non sulphide PGEM	0.21	0.04	5.66	0.00
$pn/(pn+mil)$	13.74	4.86	2.82	0.02
PGEM pred. degree of liberation	-0.54	0.13	-4.20	0.00
Regression Summary for Dependent Variable: R_s B4 excluded				
R= .95113706 R ² = .90466172 Adjusted R ² = .86890986 F(3,8)=25.304 p<.00020 Std.Error of estimate: 1.7668				
	B	St. Err. of B	t(8)	p-level
Intercept	45.39	5.16	8.80	0.00
% non sulphide PGEM	0.16	0.03	5.98	0.00
$pn/(pn+mil)$	13.49	4.15	3.25	0.01
PGEM pred. degree of liberation	-0.52	0.11	-4.69	0.00
Regression Summary for Dependent Variable: R_s C1 excluded				
R= .96326816 R ² = .92788555 Adjusted R ² = .90084263 F(3,8)=34.312 p<.00006 Std.Error of estimate: 2.1017				
	B	St. Err. of B	t(8)	p-level
Intercept	46.67	6.20	7.52	0.00
% non sulphide PGEM	0.19	0.03	6.83	0.00
$pn/(pn+mil)$	14.44	4.90	2.95	0.02
PGEM pred. degree of liberation	-0.55	0.13	-4.21	0.00
Regression Summary for Dependent Variable: R_s C2 excluded				
R= .96354958 R ² = .92842780 Adjusted R ² = .90158823 F(3,8)=34.592 p<.00006 Std.Error of estimate: 2.0978				
	B	St. Err. of B	t(8)	p-level
Intercept	46.94	6.15	7.63	0.00
% non sulphide PGEM	0.19	0.03	6.83	0.00
$pn/(pn+mil)$	14.41	4.89	2.95	0.02
PGEM pred. degree of liberation	-0.55	0.13	-4.26	0.00

Table 4b continued Regression summary for dependent variable R_s with % non-sulphide PGE mineral, $pn/(pn+mil)$ ratio and the predicted PGE mineral degree of liberation as independent variables, in turn excluding sample A5 and one other sample.

Regression Summary for Dependent Variable: R_s C3 excluded				
R= .96336193 R ² = .92806621 Adjusted R ² = .90109103 F(3,8)=34.404 p<.00006 Std.Error of estimate: 2.1021				
	B	St. Err. of B	t(8)	p-level
Intercept	46.70	6.33	7.37	0.00
% non sulphide PGEM	0.19	0.03	6.80	0.00
$pn/(pn+mil)$	14.42	4.93	2.92	0.02
PGEM pred. degree of liberation	-0.55	0.13	-4.14	0.00
Regression Summary for Dependent Variable: R_s C4 excluded				
R= .96478014 R ² = .93080071 Adjusted R ² = .90485098 F(3,8)=35.869 p<.00005 Std.Error of estimate: 2.0399				
	B	St. Err. of B	t(8)	p-level
Intercept	46.61	5.89	7.91	0.00
% non sulphide PGEM	0.19	0.03	7.06	0.00
$pn/(pn+mil)$	13.65	4.88	2.80	0.02
PGEM pred. degree of liberation	-0.54	0.13	-4.26	0.00
Regression Summary for Dependent Variable: R_s C5 excluded				
R= .96707458 R ² = .93523324 Adjusted R ² = .91094570 F(3,8)=38.507 p<.00004 Std.Error of estimate: 1.9873				
	B	St. Err. of B	t(8)	p-level
Intercept	47.92	5.86	8.17	0.00
% non sulphide PGEM	0.19	0.03	7.12	0.00
$pn/(pn+mil)$	16.14	4.95	3.26	0.01
PGEM pred. degree of liberation	-0.59	0.13	-4.59	0.00

Table 5 Regression summary for dependent variable 100-U with sample A5 excluded.

R= .80333529 R ² = .64534759 Adjusted R ² = .14883422 F(7,5)=1.2998 p<.39968 Std.Error of estimate: 1.8567				
	B	St. Err. of B	t(5)	p-level
Intercept	12.99	12.38	1.05	0.34
% non-sulphide PGE mineral	-0.03	0.04	-0.71	0.51
PGEM grain diameter at <2mm	0.12	3.43	0.04	0.97
pentlandite content	0.82	108.18	0.01	0.99
BMS grain diameter 1	0.12	0.23	0.54	0.61
pn/(pn+mil)	7.46	7.85	0.95	0.39
Predicted BMS liberation	0.08	0.08	0.99	0.37
Predicted PGEM liberation	-0.31	0.15	-2.11	0.09

Table 6 Regression summary for dependent variable k_f. All fourteen samples included.

R= .95863501 R ² = .91898107 Adjusted R ² = .64891799 F(10,3)=3.4028 p<.17100 Std.Error of estimate: .19175				
	B	St. Err. of B	t(3)	p-level
Intercept	1.76	1.23	1.44	0.25
% non-sulphide PGE mineral	0.00	0.00	0.33	0.76
PGEM grain diameter at <2mm	-0.90	0.44	-2.02	0.14
pentlandite content	-30.08	14.27	-2.11	0.13
Nickel content	0.00	0.00	0.22	0.84
PGE+Au content	0.19	0.15	1.28	0.29
Copper content	0.00	0.00	-0.02	0.98
BMS grain diameter at <2mm	-0.01	0.02	-0.62	0.58
pn/(pn+mil)	0.86	0.57	1.51	0.23
Predicted BMS liberation	0.02	0.01	2.13	0.12
Predicted PGEM liberation	0.02	0.01	1.48	0.23

Table 7 Regression summary for dependent variable k_s . All fourteen samples included.

R= .94431250 R ² = .89172610 Adjusted R ² = .53081309 F(10,3)=2.4708 p<.24710 Std.Error of estimate: .01813				
	B	St. Err. of B	t(3)	p-level
Intercept	0.19	0.12	1.63	0.20
% non-sulphide PGE mineral	0.00	0.00	0.33	0.76
PGEM grain diameter at <2mm	-0.01	0.04	-0.12	0.91
pentlandite content	1.01	1.35	0.75	0.51
Nickel content	0.00	0.00	-3.16	0.05
PGE+Au content	0.00	0.01	-0.27	0.81
Copper content	0.00	0.00	1.36	0.27
BMS grain diameter at <2mm	0.00	0.00	-1.04	0.37
pn/(pn+mil)	-0.06	0.05	-1.14	0.34
Predicted BMS liberation	0.00	0.00	0.40	0.72
Predicted PGEM liberation	0.00	0.00	1.38	0.26

Table 8a Observed and predicted values for R_f with % non-sulphide PGE mineral and the PGE mineral degree of liberation as independent variables as independent variables, in turn excluding one of the samples.

Sample number	R_f Observed	R_f Predicted (n=14)	* R_f Predicted (n=13)	R_f Observed – * R_f predicted
A2	75.1	75.4	75.5	-0.4
B2	80.3	77.5	77.0	3.3
A1	79.8	79.5	79.4	0.4
B1	73.8	69.7	69.3	4.5
A3	63.4	69.6	70.4	-7.0
B3	74.4	72.4	72.2	2.2
A4	59.9	59.2	59.0	1.0
B4	56.5	60.9	63.1	-6.6
A5	39.7	38.3	33.8	5.9
C1	66.3	72.1	73.0	-6.6
C2	71.5	71.3	71.3	0.2
C3	68.2	66.7	66.3	1.9
C4	74.8	70.6	69.9	5.0
C5	74.8	75.3	75.3	-0.5

* R_f predicted : Value of R_f determined from the regression equation calculated from the other 13 samples.

Table 8b Regression summary for dependent variable R_f with % non-sulphide PGE mineral and the PGE mineral degree of liberation as independent variables, in turn excluding one sample.

Regression Summary for Dependent Variable: R_f A2 excluded				
R= .96031566 R ² = .92220617 Adjusted R ² = .90664740 F(2,10)=59.272 p<.00000 Std.Error of estimate: 3.4096				
	B	St. Err. of B	t(10)	p-level
Intercept	40.80	5.54	7.37	0.00
PGEM degree of liberation	0.47	0.07	6.48	0.00
% non sulphide PGEM	-0.31	0.03	-9.62	0.00
Regression Summary for Dependent Variable: R_f B2 excluded				
R= .95891845 R ² = .91952459 Adjusted R ² = .90342951 F(2,10)=57.131 p<.00000 Std.Error of estimate: 3.3444				
	B	St. Err. of B	t(10)	p-level
Intercept	41.49	5.53	7.50	0.00
PGEM degree of liberation	0.46	0.07	6.20	0.00
% non sulphide PGEM	-0.30	0.03	-9.67	0.00
Regression Summary for Dependent Variable: R_f A1 excluded				
R= .95769222 R ² = .91717438 Adjusted R ² = .90060926 F(2,10)=55.368 p<.00000 Std.Error of estimate: 3.4093				
	B	St. Err. of B	t(10)	p-level
Intercept	40.50	5.80	6.99	0.00
PGEM degree of liberation	0.47	0.08	6.10	0.00
% non sulphide PGEM	-0.31	0.03	-9.61	0.00
Regression Summary for Dependent Variable: R_f B1 excluded				
R= .96636749 R ² = .93386613 Adjusted R ² = .92063935 F(2,10)=70.604 p<.00000 Std.Error of estimate: 3.1607				
	B	St. Err. of B	t(10)	p-level
Intercept	41.46	5.16	8.04	0.00
PGEM degree of liberation	0.46	0.07	6.74	0.00
% non sulphide PGEM	-0.31	0.03	-10.58	0.00
Regression Summary for Dependent Variable: R_f A3 excluded				
R= .97587086 R ² = .95232394 Adjusted R ² = .94278873 F(2,10)=99.874 p<.00000 Std.Error of estimate: 2.6865				
	B	St. Err. of B	t(10)	p-level
Intercept	43.09	4.46	9.66	0.00
PGEM degree of liberation	0.45	0.06	7.78	0.00
% non sulphide PGEM	-0.31	0.02	-12.62	0.00
Regression Summary for Dependent Variable: R_f B3 excluded				
R= .96172249 R ² = .92491014 Adjusted R ² = .90989217 F(2,10)=61.587 p<.00000 Std.Error of estimate: 3.3604				
	B	St. Err. of B	t(10)	p-level
Intercept	41.16	5.49	7.49	0.00
PGEM degree of liberation	0.46	0.07	6.39	0.00
% non sulphide PGEM	-0.31	0.03	-9.89	0.00

Table 8b continued Regression summary for dependent variable R_f with % non-sulphide PGE mineral and the PGE mineral degree of liberation as independent variables, in turn excluding one sample.

Regression Summary for Dependent Variable: R_f A4 excluded				
R= .96033113 R ² = .92223588 Adjusted R ² = .90668306 F(2,10)=59.297 p<.00000 Std.Error of estimate: 3.3729				
	B	St. Err. of B	t(10)	p-level
Intercept	41.90	5.91	7.09	0.00
PGEM degree of liberation	0.46	0.08	5.98	0.00
% non sulphide PGEM	-0.32	0.04	-8.62	0.00
Regression Summary for Dependent Variable: R_f B4 excluded				
R= .96412672 R ² = .92954033 Adjusted R ² = .91544840 F(2,10)=65.963 p<.00000 Std.Error of estimate: 3.1269				
	B	St. Err. of B	t(10)	p-level
Intercept	38.37	5.37	7.15	0.00
PGEM degree of liberation	0.50	0.07	7.15	0.00
% non sulphide PGEM	-0.29	0.03	-8.78	0.00
Regression Summary for Dependent Variable: R_f A5 excluded				
R= .90792959 R ² = .82433615 Adjusted R ² = .78920337 F(2,10)=23.463 p<.00017 Std.Error of estimate: 3.3758				
	B	St. Err. of B	t(10)	p-level
Intercept	37.67	8.54	4.41	0.00
PGEM degree of liberation	0.51	0.12	4.40	0.00
% non sulphide PGEM	-0.32	0.05	-6.85	0.00
Regression Summary for Dependent Variable: R_f C1 excluded				
R= .96778781 R ² = .93661324 Adjusted R ² = .92393589 F(2,10)=73.881 p<.00000 Std.Error of estimate: 3.1208				
	B	St. Err. of B	t(10)	p-level
Intercept	42.15	5.16	8.17	0.00
PGEM degree of liberation	0.46	0.07	6.86	0.00
% non sulphide PGEM	-0.31	0.03	-10.77	0.00
Regression Summary for Dependent Variable: R_f C2 excluded				
R= .96125836 R ² = .92401764 Adjusted R ² = .90882117 F(2,10)=60.805 p<.00000 Std.Error of estimate: 3.4116				
	B	St. Err. of B	t(10)	p-level
Intercept	40.94	5.65	7.24	0.00
PGEM degree of liberation	0.47	0.07	6.41	0.00
% non sulphide PGEM	-0.31	0.03	-9.62	0.00
Regression Summary for Dependent Variable: R_f C3 excluded				
R= .96276503 R ² = .92691650 Adjusted R ² = .91229981 F(2,10)=63.415 p<.00000 Std.Error of estimate: 3.3564				
	B	St. Err. of B	t(10)	p-level
Intercept	39.36	5.97	6.59	0.00
PGEM degree of liberation	0.49	0.08	6.37	0.00
% non sulphide PGEM	-0.30	0.03	-9.74	0.00

Table 8b continued Regression summary for dependent variable R_f with % non-sulphide PGE mineral and the PGE mineral degree of liberation as independent variables, in turn excluding one sample.

Regression Summary for Dependent Variable: R_f C4 excluded				
R= .96800335 R ² = .93703048 Adjusted R ² = .92443658 F(2,10)=74.403 p<.00000 Std.Error of estimate: 3.0710				
	B	St. Err. of B	t(10)	p-level
Intercept	38.68	5.17	7.48	0.00
PGEM degree of liberation	0.49	0.07	7.36	0.00
% non sulphide PGEM	-0.30	0.03	-10.39	0.00
Regression Summary for Dependent Variable: R_f C5 excluded				
R= .96043776 R ² = .92244070 Adjusted R ² = .90692884 F(2,10)=59.467 p<.00000 Std.Error of estimate: 3.4092				
	B	St. Err. of B	t(10)	p-level
Intercept	40.84	5.54	7.37	0.00
PGEM degree of liberation	0.47	0.07	6.49	0.00
% non sulphide PGEM	-0.31	0.03	-9.60	0.00

Table 9a Observed and predicted values for R_f with % non-sulphide PGE mineral PGE mineral degree of liberation, base-metal sulphide degree of liberation, PGE mineral grain size prior, pentlandite/(pentlandite+millerite), and the PGE mineral degree of liberation as independent variables, in turn excluding one of the samples.

Sample number	R_f Observed	R_f Predicted (n=14)	* R_f Predicted (n=13)	R_f Observed – * R_f predicted
A2	79.8	78.3	77.8	2.0
B2	75.1	73.5	72.9	2.2
A1	63.4	64.8	66.5	-3.1
B1	73.8	74.9	75.6	-1.8
A3	80.3	80.2	76.3	4.0
B3	66.3	69.5	69.3	-3.0
A4	71.5	73.1	74.0	-2.6
B4	68.2	67.8	67.7	0.5
A5	74.8	71.9	70.7	4.2
C1	74.8	73.1	70.3	4.5
C2	59.9	59.4	58.7	1.2
C3	39.7	39.0	35.0	4.6
C4	74.4	75.8	76.4	-2.0
C5	56.5	57.0	58.6	-2.0

* R_f predicted : Value of R_f determined from the regression equation calculated from the other 13 samples.

Table 9b Regression summary for dependent variable R_f with % non-sulphide PGE mineral, PGE mineral degree of liberation, base-metal sulphide degree of liberation, PGE mineral grain size prior, pentlandite/(pentlandite+millerite) to milling as independent variables, in turn excluding one sample.

Regression Summary for Dependent Variable: R_f A2 excluded				
R= .99231764 R ² = .98469429 Adjusted R ² = .97376164 F(5,7)=90.069 p<.00000 Std.Error of estimate: 1.8076				
	B	St. Err. of B	t(7)	p-level
Intercept	63.12	6.34	9.96	0.00
PGEM degree of liberation	0.48	0.04	11.00	0.00
% non-sulphide PGEM	-0.33	0.02	-13.99	0.00
PGEM grain diameter <2mm	-9.69	1.93	-5.01	0.00
BMS degree of liberation	0.08	0.03	2.67	0.03
pn/(pn+millerite)	-8.60	2.22	-3.87	0.01
Regression Summary for Dependent Variable: R_f B2 excluded				
R= .99023825 R ² = .98057179 Adjusted R ² = .96669449 F(5,7)=70.660 p<.00001 Std.Error of estimate: 1.9640				
	B	St. Err. of B	t(7)	p-level
Intercept	62.20	6.87	9.06	0.00
PGEM degree of liberation	0.47	0.05	9.97	0.00
% non-sulphide PGEM	-0.34	0.03	-12.85	0.00
PGEM grain diameter <2mm	-9.23	2.11	-4.37	0.00
BMS degree of liberation	0.08	0.03	2.38	0.05
pn/(pn+millerite)	-7.58	2.20	-3.44	0.01
Regression Summary for Dependent Variable: R_f A1 excluded				
R= .99057140 R ² = .98123170 Adjusted R ² = .96782577 F(5,7)=73.194 p<.00001 Std.Error of estimate: 1.9397				
	B	St. Err. of B	t(7)	p-level
Intercept	62.80	6.93	9.07	0.00
PGEM degree of liberation	0.47	0.05	9.74	0.00
% non-sulphide PGEM	-0.34	0.03	-13.36	0.00
PGEM grain diameter <2mm	-9.28	2.03	-4.57	0.00
BMS degree of liberation	0.08	0.03	2.42	0.05
pn/(pn+millerite)	-7.79	2.23	-3.49	0.01
Regression Summary for Dependent Variable: R_f B1 excluded				
R= .99120020 R ² = .98247784 Adjusted R ² = .96996200 F(5,7)=78.499 p<.00001 Std.Error of estimate: 1.9445				
	B	St. Err. of B	t(7)	p-level
Intercept	63.20	7.30	8.66	0.00
PGEM degree of liberation	0.47	0.05	10.18	0.00
% non-sulphide PGEM	-0.34	0.03	-13.21	0.00
PGEM grain diameter <2mm	-9.64	2.35	-4.10	0.00
BMS degree of liberation	0.08	0.03	2.43	0.05
pn/(pn+millerite)	-7.78	2.24	-3.47	0.01

Table 9b continued Regression summary for dependent variable R_f with % non-sulphide PGE mineral, PGE mineral degree of liberation, base-metal sulphide degree of liberation, PGE mineral grain size prior, pentlandite/(pentlandite+millerite) to milling as independent variables, in turn excluding one sample.

Regression Summary for Dependent Variable: R_f A3 excluded				
R= .99237798 R ² = .98481405 Adjusted R ² = .97396694 F(5,7)=90.790 p<.00000 Std.Error of estimate: 1.8122				
	B	St. Err. of B	t(7)	p-level
Intercept	60.07	6.54	9.18	0.00
PGEM degree of liberation	0.45	0.05	9.45	0.00
% non-sulphide PGEM	-0.34	0.02	-14.61	0.00
PGEM grain diameter <2mm	-7.71	2.28	-3.38	0.01
BMS degree of liberation	0.07	0.03	2.15	0.07
pn/(pn+millerite)	-5.50	2.74	-2.00	0.09
Regression Summary for Dependent Variable: R_f B3 excluded				
R= .99141211 R ² = .98289798 Adjusted R ² = .97068224 F(5,7)=80.462 p<.00000 Std.Error of estimate: 1.9168				
	B	St. Err. of B	t(7)	p-level
Intercept	63.33	6.96	9.10	0.00
PGEM degree of liberation	0.47	0.05	10.33	0.00
% non-sulphide PGEM	-0.35	0.03	-13.33	0.00
PGEM grain diameter <2mm	-9.64	2.14	-4.50	0.00
BMS degree of liberation	0.08	0.03	2.46	0.04
pn/(pn+millerite)	-7.61	2.15	-3.55	0.01
Regression Summary for Dependent Variable: R_f A4 excluded				
R= .99094658 R ² = .98197513 Adjusted R ² = .96910022 F(5,7)=76.270 p<.00001 Std.Error of estimate: 1.9409				
	B	St. Err. of B	t(7)	p-level
Intercept	62.20	6.74	9.23	0.00
PGEM degree of liberation	0.46	0.05	8.81	0.00
% non-sulphide PGEM	-0.35	0.03	-12.41	0.00
PGEM grain diameter <2mm	-9.00	2.04	-4.41	0.00
BMS degree of liberation	0.08	0.03	2.45	0.04
pn/(pn+millerite)	-7.34	2.23	-3.29	0.01
Regression Summary for Dependent Variable: R_f B4 excluded				
R= .99025840 R ² = .98061170 Adjusted R ² = .96676291 F(5,7)=70.808 p<.00001 Std.Error of estimate: 1.9605				
	B	St. Err. of B	t(7)	p-level
Intercept	60.25	10.99	5.48	0.00
PGEM degree of liberation	0.48	0.05	8.84	0.00
% non-sulphide PGEM	-0.33	0.04	-7.84	0.00
PGEM grain diameter <2mm	-8.73	2.83	-3.09	0.02
BMS degree of liberation	0.08	0.04	2.28	0.06
pn/(pn+millerite)	-7.56	2.19	-3.45	0.01

Table 9b continued Regression summary for dependent variable R_f with % non-sulphide PGE mineral, PGE mineral degree of liberation, base-metal sulphide degree of liberation, PGE mineral grain size prior, pentlandite/(pentlandite+millerite) to milling as independent variables, in turn excluding one sample.

Regression Summary for Dependent Variable: R_f A5 excluded				
R= .97920845 R ² = .95884918 Adjusted R ² = .92945574 F(5,7)=32.621 p<.00010 Std.Error of estimate: 1.9529				
	B	St. Err. of B	t(7)	p-level
Intercept	59.42	10.81	5.50	0.00
PGEM degree of liberation	0.50	0.10	5.21	0.00
% non-sulphide PGEM	-0.34	0.03	-12.26	0.00
PGEM grain diameter <2mm	-8.91	2.16	-4.13	0.00
BMS degree of liberation	0.08	0.03	2.38	0.05
pn/(pn+millerite)	-7.88	2.40	-3.28	0.01
Regression Summary for Dependent Variable: R_f C1 excluded				
R= .99316814 R ² = .98638296 Adjusted R ² = .97665651 F(5,7)=101.41 p<.00000 Std.Error of estimate: 1.7289				
	B	St. Err. of B	t(7)	p-level
Intercept	62.23	6.00	10.37	0.00
PGEM degree of liberation	0.48	0.04	11.52	0.00
% non-sulphide PGEM	-0.34	0.02	-15.26	0.00
PGEM grain diameter <2mm	-8.67	1.82	-4.75	0.00
BMS degree of liberation	0.06	0.03	1.94	0.09
pn/(pn+millerite)	-7.64	1.94	-3.95	0.01
Regression Summary for Dependent Variable: R_f C2 excluded				
R= .99406352 R ² = .98816227 Adjusted R ² = .97970675 F(5,7)=116.87 p<.00000 Std.Error of estimate: 1.6095				
	B	St. Err. of B	t(7)	p-level
Intercept	61.87	5.59	11.07	0.00
PGEM degree of liberation	0.47	0.04	12.31	0.00
% non-sulphide PGEM	-0.34	0.02	-16.12	0.00
PGEM grain diameter <2mm	-9.38	1.67	-5.60	0.00
BMS degree of liberation	0.10	0.03	3.42	0.01
pn/(pn+millerite)	-9.31	2.03	-4.58	0.00
Regression Summary for Dependent Variable: R_f C3 excluded				
R= .99140382 R ² = .98288153 Adjusted R ² = .97065406 F(5,7)=80.383 p<.00000 Std.Error of estimate: 1.9415				
	B	St. Err. of B	t(7)	p-level
Intercept	61.79	6.77	9.12	0.00
PGEM degree of liberation	0.48	0.05	9.85	0.00
% non-sulphide PGEM	-0.34	0.02	-13.64	0.00
PGEM grain diameter <2mm	-9.15	2.01	-4.55	0.00
BMS degree of liberation	0.07	0.03	2.14	0.07
pn/(pn+millerite)	-7.32	2.24	-3.26	0.01



Table 9b continued Regression summary for dependent variable R_f with % non-sulphide PGE mineral, PGE mineral degree of liberation, base-metal sulphide degree of liberation, PGE mineral grain size prior, pentlandite/(pentlandite+millerite) to milling as independent variables, in turn excluding one sample.

Regression Summary for Dependent Variable: R_f C4 excluded				
R= .99504397 R ² = .99011250 Adjusted R ² = .98305001				
F(5,7)=140.19 p<.00000 Std.Error of estimate: 1.4545				
	B	St. Err. of B	t(7)	p-level
Intercept	61.62	5.05	12.20	0.00
PGEM degree of liberation	0.48	0.03	13.72	0.00
% non-sulphide PGEM	-0.34	0.02	-18.33	0.00
PGEM grain diameter <2mm	-9.00	1.51	-5.96	0.00
BMS degree of liberation	0.06	0.03	2.32	0.05
pn/(pn+millerite)	-6.01	1.75	-3.43	0.01
Regression Summary for Dependent Variable: R_f C5 excluded				
R= .99169877 R ² = .98346645 Adjusted R ² = .97165678				
F(5,7)=83.276 p<.00000 Std.Error of estimate: 1.8813				
	B	St. Err. of B	t(7)	p-level
Intercept	60.24	6.92	8.71	0.00
PGEM degree of liberation	0.46	0.05	9.66	0.00
% non-sulphide PGEM	-0.33	0.03	-12.93	0.00
PGEM grain diameter <2mm	-9.03	1.96	-4.62	0.00
BMS degree of liberation	0.11	0.05	2.31	0.05
pn/(pn+millerite)	-7.65	2.11	-3.63	0.01

Table 10a Observed and predicted values for variable R_s with % non-sulphide PGE mineral, PGE mineral degree of liberation, chromite grain diameter and PGE mineral diameter prior to milling, as independent variables, in turn excluding one of the samples.

Sample number	R_s Observed	R_s Predicted (n=14)	* R_s Predicted (n=13)	R_s Observed – * R_s predicted
A2	16.3	17.5	18.3	-1.2
B2	18.8	18.4	18.4	0.4
A1	25.9	25.9	26.0	-0.1
B1	19.6	19.5	19.5	0.1
A3	15.3	15.6	15.7	-0.3
B3	24.5	24.4	24.4	0.2
A4	22.6	23.0	23.1	-0.5
B4	24.0	24.8	24.9	-0.9
A5	19.9	18.9	18.3	1.0
C1	21.1	20.7	20.6	0.4
C2	33.7	34.0	34.0	-0.3
C3	28.7	28.9	30.8	-0.2
C4	21.0	20.2	19.8	0.8
C5	37.7	37.3	36.4	0.4

* R_s predicted : Value of R_s determined from the regression equation calculated for the other 13 samples.

Table 10b Regression summary for dependent variable R_s with % non-sulphide PGE mineral, PGE mineral degree of liberation, chromite grain diameter and PGE mineral diameter prior to milling as independent variables, in turn excluding one sample.

Regression Summary for Dependent Variable: R_s A2 excluded				
$R = .99640518$ $R^2 = .99282328$ Adjusted $R^2 = .98923491$				
$F(4,8) = 276.68$ $p < .00000$ Std. Error of estimate: .66807				
	B	St. Err. of B	t(8)	p-level
Intercept	-6.02	2.14	-2.81	0.02
PGEM degree of liberation	-0.16	0.02	-7.29	0.00
% non-sulphide PGEM	0.20	0.01	19.97	0.00
PGEM ECD <2mm	5.98	0.69	8.65	0.00
Chromite ECD <2mm	0.13	0.01	10.32	0.00
Regression Summary for Dependent Variable: R_s B2 excluded				
$R = .99594658$ $R^2 = .99190959$ Adjusted $R^2 = .98786439$				
$F(4,8) = 245.21$ $p < .00000$ Std. Error of estimate: .67398				
	B	St. Err. of B	t(8)	p-level
Intercept	-5.94	2.19	-2.71	0.03
PGEM degree of liberation	-0.16	0.02	-6.82	0.00
% non-sulphide PGEM	0.20	0.01	19.52	0.00
PGEM ECD <2mm	5.99	0.70	8.55	0.00
Chromite ECD <2mm	0.12	0.01	9.95	0.00

Table 10b continued Regression summary for dependent variable R_s with % non-sulphide PGE mineral, PGE mineral degree of liberation, chromite grain diameter and PGE mineral diameter prior to milling as independent variables, in turn excluding one sample.

Regression Summary for Dependent Variable: R_s A1 excluded				
$R = .99638789$ $R^2 = .99278883$ $Adjusted R^2 = .98918324$ $F(4,8) = 275.35$ $p < .00000$ $Std. Error of estimate: .66609$				
	B	St. Err. of B	t(8)	p-level
Intercept	-3.96	4.43	-0.90	0.40
PGEM degree of liberation	-0.17	0.03	-6.06	0.00
% non-sulphide PGEM	0.21	0.02	12.39	0.00
PGEM ECD <2mm	5.94	0.70	8.46	0.00
Chromite ECD <2mm	0.12	0.02	6.61	0.00
Regression Summary for Dependent Variable: R_s B1 excluded				
$R = .99636838$ $R^2 = .99274995$ $Adjusted R^2 = .98912492$ $F(4,8) = 273.86$ $p < .00000$ $Std. Error of estimate: .67678$				
	B	St. Err. of B	t(8)	p-level
Intercept	-5.88	2.38	-2.47	0.04
PGEM degree of liberation	-0.16	0.02	-7.22	0.00
% non-sulphide PGEM	0.20	0.01	18.25	0.00
PGEM ECD <2mm	5.95	0.81	7.38	0.00
Chromite ECD <2mm	0.13	0.01	10.19	0.00
Regression Summary for Dependent Variable: R_s A3 excluded				
$R = .99640518$ $R^2 = .99282328$ $Adjusted R^2 = .98923491$ $F(4,8) = 276.68$ $p < .00000$ $Std. Error of estimate: .66807$				
	B	St. Err. of B	t(8)	p-level
Intercept	-6.02	2.14	-2.81	0.02
PGEM degree of liberation	-0.16	0.02	-7.29	0.00
% non-sulphide PGEM	0.20	0.01	19.97	0.00
PGEM ECD <2mm	5.98	0.69	8.65	0.00
Chromite ECD <2mm	0.13	0.01	10.32	0.00
Regression Summary for Dependent Variable: R_s A4 excluded				
$R = .99571841$ $R^2 = .99145516$ $Adjusted R^2 = .98718274$ $F(4,8) = 232.06$ $p < .00000$ $Std. Error of estimate: .66131$				
	B	St. Err. of B	t(8)	p-level
Intercept	-6.29	2.16	-2.91	0.02
PGEM degree of liberation	-0.16	0.02	-7.42	0.00
% non-sulphide PGEM	0.20	0.01	19.86	0.00
PGEM ECD <2mm	5.94	0.69	8.61	0.00
Chromite ECD <2mm	0.13	0.01	10.24	0.00

Table 10 continued Regression summary for dependent variable R_s with % non-sulphide PGE mineral, PGE mineral degree of liberation, chromite grain diameter and PGE mineral diameter prior to milling as independent variables, in turn excluding one sample.

Regression Summary for Dependent Variable: R_s B3 excluded				
$R = .99714580$ $R^2 = .99429975$ $Adjusted R^2 = .99144963$				
$F(4,8) = 348.86$ $p < .00000$ $Std. Error of estimate: .60564$				
	B	St. Err. of B	t(8)	p-level
Intercept	-7.12	2.09	-3.41	0.01
PGEM degree of liberation	-0.16	0.02	-7.56	0.00
% non-sulphide PGEM	0.21	0.01	20.22	0.00
PGEM ECD <2mm	6.56	0.73	8.96	0.00
Chromite ECD <2mm	0.12	0.01	10.31	0.00
Regression Summary for Dependent Variable: R_s B4 excluded				
$R = .99485165$ $R^2 = .98972981$ $Adjusted R^2 = .98459472$				
$F(4,8) = 192.74$ $p < .00000$ $Std. Error of estimate: .62593$				
	B	St. Err. of B	t(8)	p-level
Intercept	-3.48	2.96	-1.18	0.27
PGEM degree of liberation	-0.18	0.02	-7.48	0.00
% non-sulphide PGEM	0.19	0.01	13.98	0.00
PGEM ECD <2mm	5.36	0.85	6.29	0.00
Chromite ECD <2mm	0.13	0.01	11.23	0.00
Regression Summary for Dependent Variable: R_s A5 excluded				
$R = .99638789$ $R^2 = .99278883$ $Adjusted R^2 = .98918324$				
$F(4,8) = 275.35$ $p < .00000$ $Std. Error of estimate: .66609$				
	B	St. Err. of B	t(8)	p-level
Intercept	-3.96	4.43	-0.90	0.40
PGEM degree of liberation	-0.17	0.03	-6.06	0.00
% non-sulphide PGEM	0.21	0.02	12.39	0.00
PGEM ECD <2mm	5.94	0.70	8.46	0.00
Chromite ECD <2mm	0.12	0.02	6.61	0.00
Regression Summary for Dependent Variable: R_s C1 excluded				
$R = .99699872$ $R^2 = .99400644$ $Adjusted R^2 = .99100967$				
$F(4,8) = 331.69$ $p < .00000$ $Std. Error of estimate: .62448$				
	B	St. Err. of B	t(8)	p-level
Intercept	-6.36	2.02	-3.14	0.01
PGEM degree of liberation	-0.17	0.02	-7.95	0.00
% non-sulphide PGEM	0.20	0.01	21.34	0.00
PGEM ECD <2mm	6.14	0.65	9.43	0.00
Chromite ECD <2mm	0.13	0.01	11.24	0.00

Table 10 continued Regression summary for dependent variable R_s with % non-sulphide PGE mineral, PGE mineral degree of liberation, chromite grain diameter and PGE mineral diameter prior to milling as independent variables, in turn excluding one sample.

Regression Summary for Dependent Variable: R_s C2 excluded				
$R = .99646806$ $R^2 = .99294860$ $Adjusted R^2 = .98942290$ $F(4,8) = 281.63$ $p < .00000$ $Std. Error of estimate: .67745$				
	B	St. Err. of B	t(8)	p-level
Intercept	-6.03	2.18	-2.77	0.02
PGEM degree of liberation	-0.16	0.02	-7.21	0.00
% non-sulphide PGEM	0.20	0.01	19.68	0.00
PGEM ECD <2mm	6.01	0.70	8.55	0.00
Chromite ECD <2mm	0.12	0.01	10.30	0.00
Regression Summary for Dependent Variable: R_s C3 excluded				
$R = .99730367$ $R^2 = .99461461$ $Adjusted R^2 = .99192191$ $F(4,8) = 369.38$ $p < .00000$ $Std. Error of estimate: .59246$				
	B	St. Err. of B	t(8)	p-level
Intercept	-5.49	1.93	-2.84	0.02
PGEM degree of liberation	-0.18	0.02	-8.02	0.00
% non-sulphide PGEM	0.20	0.01	21.36	0.00
PGEM ECD <2mm	5.90	0.61	9.61	0.00
Chromite ECD <2mm	0.13	0.01	11.50	0.00
Regression Summary for Dependent Variable: R_s C4 excluded				
$R = .99716555$ $R^2 = .99433913$ $Adjusted R^2 = .99150870$ $F(4,8) = 351.30$ $p < .00000$ $Std. Error of estimate: .59931$				
	B	St. Err. of B	t(8)	p-level
Intercept	-5.87	1.93	-3.05	0.02
PGEM degree of liberation	-0.17	0.02	-8.30	0.00
% non-sulphide PGEM	0.20	0.01	21.05	0.00
PGEM ECD <2mm	5.61	0.68	8.31	0.00
Chromite ECD <2mm	0.13	0.01	11.03	0.00
Regression Summary for Dependent Variable: R_s C5 excluded				
$R = .99668489$ $R^2 = .99338078$ $Adjusted R^2 = .99007116$ $F(4,8) = 300.15$ $p < .00000$ $Std. Error of estimate: .65301$				
	B	St. Err. of B	t(8)	p-level
Intercept	-6.01	2.10	-2.87	0.02
PGEM degree of liberation	-0.16	0.02	-7.52	0.00
% non-sulphide PGEM	0.20	0.01	20.37	0.00
PGEM ECD <2mm	6.01	0.67	8.94	0.00
Chromite ECD <2mm	0.12	0.01	10.66	0.00



Table 11 Regression summary for dependent variable 100-U excluding sample A5.

R= .88368204 R ² = .78089395 Adjusted R ² = .56178790 F(6,6)=3.5640 p<.07365 Std.Error of estimate: 1.3322				
	B	St. Err. of B	t(6)	p-level
Intercept	21.42	6.50	3.29	0.02
PGEM degree of liberation	-0.33	0.08	-3.98	0.01
% non-sulphide PGEM	0.06	0.03	2.35	0.06
PGEM ECD 80%<75µm	3.63	3.51	1.03	0.34
BMS degree of liberation	-0.05	0.03	-1.75	0.13
BMS ECD 80%<75µm	0.25	0.19	1.30	0.24
pn/(pn+mil)	5.61	2.38	2.36	0.06

Table 12 Regression summary for dependent variable k_f. All fourteen samples included.

R= .92487051 R ² = .85538547 Adjusted R ² = ----- F(12,1)=.49291 p<.82017 Std.Error of estimate: .44372				
	B	St. Err. of B	t(1)	p-level
Intercept	1.39	8.83	0.16	0.90
PGEM degree of liberation	0.01	0.03	0.48	0.72
% non-sulphide PGEM	0.00	0.01	-0.09	0.95
PGEM ECD 80%<75µm	0.51	1.20	0.42	0.74
pentlandite content	-20.52	31.24	-0.66	0.63
PGEM/(PGEM+BMS)	-2.28	6.69	-0.34	0.79
Chromite ECD <2mm	0.00	0.03	0.02	0.98
BMS degree of liberation	0.00	0.01	0.24	0.85
BMS ECD 80%<75µm	-0.03	0.10	-0.29	0.82
pn/(pn+mil)	0.81	1.67	0.48	0.71

Table 13 Regression summary for dependent variable k_s. All fourteen samples included.

R= .93283790 R ² = .87018655 Adjusted R ² = .57810628 F(9,4)=2.9793 p<.15267 Std.Error of estimate: .01719				
	B	St. Err. of B	t(4)	p-level
Intercept	0.44	0.19	2.38	0.08
PGEM degree of liberation	0.00	0.00	0.27	0.80
% non-sulphide PGEM	0.00	0.00	-1.04	0.36
PGEM ECD 80%<75µm	-0.01	0.04	-0.20	0.85
pentlandite content	0.58	0.88	0.66	0.55
PGEM/(PGEM+BMS)	-0.20	0.11	-1.75	0.16
Chromite ECD <2mm	0.00	0.00	0.44	0.68
BMS degree of liberation	0.00	0.00	-0.25	0.81
BMS ECD 80%<75µm	-0.01	0.00	-2.44	0.07
pn/(pn+mil)	-0.06	0.06	-0.99	0.38