

Article

# Magmatic PGE Sulphide Mineralization in Clinopyroxenite from the Platreef, Bushveld Complex, South Africa

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Received: 25 May 2020; Accepted: 23 June 2020; Published: 25 June 2020



**Abstract:** The Platreef, at the base of the northern limb of the Bushveld Complex in South Africa, hosts platinum-group element (PGE) mineralization in association with base-metal sulphides (BMS) and platinum-group minerals (PGM). However, whilst a magmatic origin of the stratiform mineralization of the upper Platreef has been widely confirmed, the processes responsible for the PGE and BMS mineralization and metasomatism of the host rocks in the Platreef are still under discussion. In order to contribute to the present discussion, we present an integrated petrographical, mineral-chemical, whole-rock trace- and major-element, sulphur- and neodymium-isotope, study of Platreef footwall clinopyroxenite drill core samples from Overysel, which is located in the northern sector of the northern Bushveld limb. A metasomatic transformation of magmatic pyroxenite units to non-magmatic clinopyroxenite is in accordance with the petrography and whole-rock chemical analysis. The whole-rock data display lower SiO<sub>2</sub>, FeO, Na<sub>2</sub>O and Cr (<1700 ppm), and higher CaO, concentrations in the here-studied footwall Platreef clinopyroxenite samples than primary magmatic Platreef pyroxenite and norite. The presence of capped globular sulphides in some samples, which display differentiation into pyrrhotite and pentlandite in the lower, and chalcopyrite in the upper part, is attributed to the fractional crystallization of a sulphide liquid, and a downward transport of the blebs. In situ sulphur (V-CDT) isotope BMS data show isotopic signatures ( $\delta^{34}\text{S} = 0.9$  to  $3.1$  ‰;  $\Delta^{33}\text{S} = 0.09$  to  $0.32$  ‰) close to or within the pristine magmatic range. Elevated (non-zero)  $\Delta^{33}\text{S}$  values are common for Bushveld magmas, indicating contamination by older, presumably crustal sulphur in an early stage chamber, whereas magmatic  $\delta^{34}\text{S}$  values suggest the absence of local crustal contamination during emplacement. This is in accordance with the  $\epsilon\text{Nd}$  (2.06 Ga) (chondritic uniform reservoir (CHUR)) values, of  $-6.16$  to  $-6.94$ , which are similar to those of the magmatic pyroxenite and norite of the Main Zone and the Platreef in the northern sector of the northern Bushveld limb. Base-metal sulphide textures and S–Se-ratios give evidence for a secondary S-loss during late- to post-magmatic hydrothermal alteration. The textural evidence, as well as the bulk S/Se ratios and sulphide S isotopes studies, suggest that the mineralization in both the less and the pervasively hydrothermally altered clinopyroxenite samples of Overysel are of magmatic origin. This is further supported by the PPGE (Rh, Pt, Pd) concentrations in the BMS and mass-balance calculations, in both of which large proportions of the whole-rock Pd and Rh are hosted by pentlandite, whereas Pt and

the IPGE (Os, Ir, Ru) were interpreted to mainly occur in discrete PGM. However, the presence of pentlandite with variable PGE concentrations on the thin section scale may be related to variations in the S content, already at S-saturation during magmatic formation, and/or post-solidification mobilization and redistribution.

**Keywords:** Bushveld Complex; Platreef; clinopyroxenite; fluid–rock interaction; metasomatism

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## 1. Introduction

The PGE-mineralization of the Platreef is thought to be much more diverse and complex than that of the other two major PGE deposits in the Bushveld Complex of South Africa, the narrow stratabound Merensky reef and the UG-2 chromitite (e.g., [1,2]). For instance, recent studies revealed that the syn- and post-solidification of fluids, and the country-rock assimilation, played a decisive role in concerning the PGE-mineralization along the strike [3–6]. Furthermore, the mineralizations at the top and the base of the Platreef are inconsistent in grade and thickness along the strike. Nevertheless, the PGE concentrations hosted by base-metal sulphides (BMS) are in general agreement with those of the Merensky Reef and the UG-2 [7–14]. Large proportions of the whole-rock Pd and Rh are hosted by pentlandite, while Pt and the IPGE (Ir, Ru, Os) mainly occur in platinum-group minerals (PGM), rather than being hosted by BMS. However, when comparing the Platreef's pentlandite contents to those of the Merensky reef and the UG-2, the average Pd and Rh concentrations are distinctly lower in the Platreef [9,10,12–14].

Pentlandite, pyrrhotite and chalcopyrite mostly occur as disseminated interstitial blebs, but in places also occur as massive sulphides or net-textured ores in the Platreef [5,15]. At the top of the Platreef, a magmatic origin of the stratiform mineralization, similar to that of the Merensky reef, seems to be confirmed by pyroxene mineral-chemistry, Re-Os and S-isotope studies (e.g., [12,16–19]). The mineralization of the footwall of the Platreef, which displays variable degrees of the incorporation, assimilation and reaction of the footwall lithologies with the basic magma, and pervasive hydrothermal alteration, also shows different degrees of redistribution and S assimilation [19–27].

Sporadic mineralization ranging, over tens of meters, also occurs in footwall lithologies such as clinopyroxenite (also called diopsidic clinopyroxenite or parapyroxenite) and serpentinite [15,28]. The Platreef clinopyroxenites are diopside-rich calcsilicate rocks, which were interpreted to be of metamorphic/metasomatic origin [3,6,20]. The term clinopyroxenite is used throughout the paper for calcsilicate rocks of the drill core studied here.

Within the upper, unaltered sections of the Platreef, the PGE mineralization shows a close spatial association with the BMS, while conversely, within pervasively altered footwall rocks, the fluid–rock interaction was interpreted to display decoupling of the PGE from the BMS. Thus, low BMS contents associated with high PGE grades, and high BMS contents with low PGE grades, are currently explained by late-stage hydrothermal mobilization and redistribution of PGE (and/or BMS) throughout the footwall of the Platreef (e.g., [5,9] cf. [12,20,29]).

This investigation aims to provide new observations on the distribution and controls of PGE sulphide mineralization, developed within the footwall clinopyroxenite of the Platreef at Overysel, by using petrographical, mineral-chemical, whole-rock trace- and major-element, and sulphur- and neodymium-isotope, data. In addition, the findings are complemented by quantification of the composition of the associated BMS and their textural behaviour during these processes, which, in particular, has implications for the future exploration potential of the Platreef footwall rocks in the investigated area.

## 2. Geological Setting

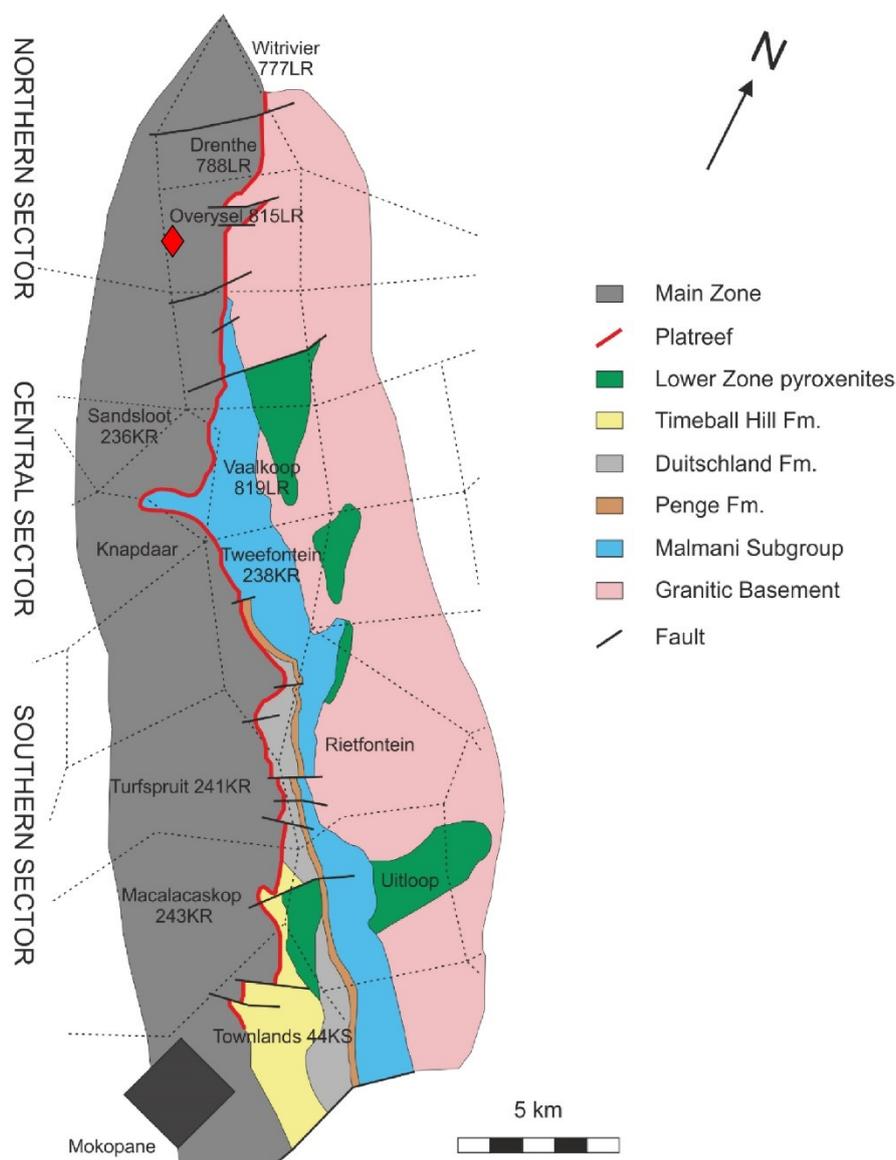
The 2.054 Ga Bushveld Complex in South Africa represents the largest layered igneous complex in the world, covering an area >66,000 km<sup>2</sup>, and varying in thickness between 7 and 8 km [30–34]. The Bushveld Complex, which is located in the northern Kaapvaal Craton, consists of five limbs, namely, the western and eastern limbs, which are roughly symmetric, a largely eroded far western limb, a sediment-covered southern (Bethal) limb, and the northern limb [35]. The Bushveld Complex is further subdivided into the mafic rocks of the Rustenburg Layered Suite (RLS), the Lebowa Granite Suite, the Raseebie Granophyre Suite and the siliceous rocks of the Rooiberg Group [36,37].

The Bushveld Complex contains the world's largest and economically most important resource of platinum-group elements (PGE), as well as other important elements, such as chromium, vanadium, copper and fluorine [30,38–40]. The economic PGE mineralization, which is hosted within the RLS, mainly occurs in three ore bodies; the thin (ca. 1 m thick) but extensive (>100 km strike length) Merensky reef, the UG-2 chromitite (with 5 to 8 ppm PGE), and the thicker (>30 m thick) Platreef, with 3 to 4 ppm PGE, which comprise the world's three largest PGE ore horizons [35,41,42].

The RLS is usually subdivided into the Marginal, Lower, Critical, Main and Upper Zones, and the Basal Ultramafic Sequence [43]. This subdivision is also applied to the northern limb, but differs in several aspects. For instance, the Platreef was interpreted to represent a northern facies of the Merensky reef, or an analogue of the Critical Zone (e.g., [18,22,44–48]). Furthermore, it was suggested that the emplacement of the Main Zone magma, which is thought to have supplied Ni, Cu and PGE to the Merensky reef and the UG-2, in the northern Bushveld limb significantly post-dates the Platreef, and thus cannot have contributed Ni, Cu and PGE to the Platreef [49,50]. This seems to be in contrast with the eastern and western limbs of the Bushveld, where the Main Zone magma is thought to have been involved in the formation of the Merensky reef [51–55].

The northern limb of the Bushveld Complex is situated around 300 km NNW of Johannesburg. The 100-km long and 15-km wide sinuous outcrop (Figure 1) strikes N–S and dips ~WSW. The layered mafic rocks of the Platreef overlie a succession of progressively older sedimentary suites of the Transvaal Supergroup and Archean basement, comprising basement granites and gneisses [20,48,56].

The thickness of the Platreef varies, from 30 m up to 400 m [15,22]. It is interpreted to represent a sill or complex that intruded as a crystal mush into the basement granite–gneiss and sediments of the Transvaal Supergroup (e.g., [28]). The Platreef is further subdivided based on the variation of the footwall lithologies along the strike (Figure 1): (a) a northern sector underlain by an Archean granite basement on the Overysel and Drenthe farms; (b) a central sector with Malmani Dolomite as footwall on the Zwartfontein, Sandsloot, Vaalkop and Tweefontein farms; and (c) a southern sector comprising Transvaal shale, banded ironstone, mudstone and siltstone as footwall rocks, reaching from Tweefontein Hill to Townlands [22]. Recently, several Main Zone Ni–Cu–PGE magmatic sulphide deposits, like Aurora, Waterberg and Moordrift, were discovered in the northern Bushveld limb [29,51,54,57–60]. One of the most significant discoveries has been the down-dip extension of the Platreef (called Flatreef) at Turfspruit, which was interpreted to be related to the Critical Zone of the western and eastern limbs of the Bushveld Complex [44].



**Figure 1.** Geological map of the northern limb of the Bushveld Complex, showing the Platreef north of Mokopane modified after [61]. The borehole location (OY541) is indicated by the red diamond.

Detailed descriptions of the Platreef lithologies are given in Armitage, McDonald, Edwards and Manby [15]; Harris and Chaumba [3]; Kinnaird, Hutchinson, Schurmann, Nex and de Lange [22]; McDonald, Holwell and Armitage [62]; Holwell, McDonald and Armitage [20]; Kinnaird, Yudovskaya, McCreech, Huthmann and Botha [58]; McDonald, Harmer, Holwell, Hughes and Boyce [54] and Grobler, Brits, Maier and Crossingham [44]. The footwall lithologies beneath the primary igneous Platreef at Overysel comprise Archean gneiss, banded granulite (local term: granofelses), footwall clinopyroxenite (historically called parapyroxenite), calcsilicate rocks, dolomite and serpentinite. The Platreef contains >4 g/t PGE where the footwall is made of Malmani Dolomite, and 1 to 2 g/t elsewhere, with sporadically elevated values up to 10 g/t [5]. The clinopyroxenites are described to occur in the footwall of the Platreef below the igneous pyroxenites and above the Malmani Dolomite, and comprise diopside and bulk-rock Cr contents of <2000 ppm, while the igneous pyroxenite mainly consists of orthopyroxene and cumulus Ca-poorer clinopyroxene, and contains bulk-rock Cr contents of > 2000 ppm, according to the definitions of Harris and Chaumba [3], Armitage, McDonald, Edwards and Manby [15], and Holwell, McDonald and Armitage [20]. The actual calcsilicate rocks or xenoliths within the Platreef, however, are described as a mixture of dolomite and igneous reef pyroxenites, and have lower SiO<sub>2</sub>

(<40 wt %), lower FeO/MgO, and (much) higher CaO/MgO than the non-igneous clinopyroxenites [3]. Most of these lithologies underwent, or were the product of, thermal metamorphism and hydrothermal alteration, as well as sediment assimilation, which was thought to have led to a significant change in the bulk-rock chemistry and thus mineralogy of the footwall rock types. However, the exact cause of the post-magmatic processes, and the impact on the primary igneous mineralization, is still a matter of discussion [3,5,15,18,19,49]. In order to understand the nature of these processes and their influence on the sulphide-associated PGE mineralization of the clinopyroxenites, the present study is focused on characterizing the mineralization within drill core clinopyroxenite samples from Overysel.

### 3. Samples and Analytical Methods

A total of 54 samples for this study were collected from two drill cores of OY541 at Overysel 815LR (Anglo American Platinum), with corresponding coordinates referring to the WGS84 (South African Grid) Lo.29 grid (X: 50187.18; Y: 13091.78). Drill core OY541 was chosen for the present study, because it covers a typical section of clinopyroxenite between the magmatic Platreef and Malmani Dolomite (Figure 1) [6]. Preferably high-grade samples were chosen for the present study.

A total of 16 representative clinopyroxenite drill core samples were analyzed for bulk-rock Pt, Pd, Au and S, as well as major and trace-element contents (Table 1). All analyses were done by ACTLABS (Canada), using fusion coupled ICP-MS and combined fire assay ICP-MS methods. Major elements were determined with analytical uncertainties of less than 5%. Trace elements were analyzed with analytical uncertainties of mostly less than 10%.

Scanning electron microscope (SEM) investigations, which were primarily used for the detection and semi-quantitative analyses of PGM, were undertaken at the Bundesanstalt für Geowissenschaften und Rohstoffe (BGR) in Hannover, Germany, using a MLA 650F SEM (FEI Company, Hillsboro, OR, USA).

Quantitative mineral chemical analyses of sulphides and pyroxenes (Tables 2 and 3) were carried out on three polished thin sections and five polished ore sections, using a JEOL JXN-8200 electron microprobe (EMP) at the GeoZentrum Nordbayern, University of Erlangen. The JEOL JXN-8200 was operated with an acceleration voltage of 20 kV, a 3- $\mu$ m beam diameter and a probe current of 20 nA (for further analytical details see [13]).

Neodymium isotope (CHUR) measurements and Sm and Nd concentrations were determined in Class 1000 metal-free laboratories at the GeoZentrum Nordbayern. For Nd isotope measurements (Table 4), approximately 100 mg of sample powder was digested in 0.5 mL 15 M HNO<sub>3</sub> and 2 mL 12 M HF in sealed Teflon beakers overnight at 80 °C. A quantity of 0.2 mL of HClO<sub>4</sub> was then added to the sample solution, which was evaporated to near dryness at 130 °C. The sample was taken up in 2 mL 15 M HNO<sub>3</sub>, and evaporated to complete dryness at 150 °C. This step was repeated 3 times, until the sample was completely in solution, before redissolving the sample in 2 mL 3.5 M HNO<sub>3</sub>. The rare-earth elements were separated from the rock matrix using 0.2 mL of Eichrom TRU-spec resin. The sample in 3.5 M HNO<sub>3</sub> was loaded onto the resin, which was washed 3 times with 2 mL 3.5 M HNO<sub>3</sub> to remove major elements, and the rare earth elements (REE) were selectively eluted with 2 mL 2.5 M HCl. This solution was evaporated, the residue redissolved in 0.5 mL 0.25 M HCl, and Nd was then separated from the other REE using 1.5 mL Eichrom LN-spec resin in 0.25 M HCl. All reagents used were Teflon distilled, and the Nd blank was below 20 pg.

Nd isotope measurements were carried out using a Thermo Triton thermal ionization mass spectrometer in static mode. Nd was loaded in 3  $\mu$ L 0.1 M H<sub>3</sub>PO<sub>4</sub> onto the Ta filament of a double Ta-Re filament assembly and analyzed as the metal. Interference of <sup>144</sup>Sm on <sup>144</sup>Nd was corrected for by measuring <sup>147</sup>Sm, but was negligible for all samples. Instrumental mass fractionation was corrected for, assuming a <sup>144</sup>Nd/<sup>146</sup>Nd ratio of 0.7219. An in-house Nd standard yielded <sup>143</sup>Nd/<sup>144</sup>Nd = 0.511541  $\pm$  0.000008 (*n* = 6), equivalent to 0.511852 for the La Jolla Nd standard.

**Table 1.** Bulk rock compositions of the clinopyroxenite samples from drill core OY541.

Sample ID		OY541-1	OY541-2	OY541-3a1	OY541-3a2	OY541-3b	OY541-4a1	OY541-4a2	OY541-4b1	OY541-4b2	OY541-5	OY541-6a	OY541-6b	OY541-7a	OY541-7b	OY541-7c	OY541-8
Depth top (m)		1124.41	1127.02	1131.105			1132.135				1133.975	1137.04		1139.56			1142.01
Depth bottom (m)		1124.65	1127.21	1131.44			1132.425				1133.11	1137.49		1139.13			1142.19
SiO <sub>2</sub>	(wt %)	43.7	50.3	55.6	50.1	47.3	45.2	54.8	53.3	46.4	42.6	41.3	42.6	32.1	31.9	42.6	39.8
TiO <sub>2</sub>	(wt %)	0.65	0.13	0.05	0.05	0.17	0.16	0.05	0.13	0.15	0.03	0.32	0.42	0.31	0.32	0.05	0.2
Al <sub>2</sub> O <sub>3</sub>	(wt %)	1.32	4.73	0.57	0.76	4.54	4.16	1.13	1.32	4.73	17.6	10	8.88	18.5	18.5	3.59	1.32
FeO	(wt %)	11	5.55	2.32	2.71	12.9	7.48	1.42	3.48	5.93	1.42	6.06	5.03	2.32	2.71	4.13	3.99
MgO	(wt %)	30.7	19.9	17.8	15.8	23.7	20.6	18.6	17.4	21.4	3.32	18.8	21.6	27.2	29.2	24.7	33.5
CaO	(wt %)	3.78	17.2	25.6	28.1	7.28	16.8	25.2	24.4	18.1	34.9	18.2	17.2	10.5	6.72	16.2	10.4
K <sub>2</sub> O	(wt %)	<0.10	<0.10	<0.10	<0.10	<0.10	<0.10	<0.10	<0.10	<0.10	<0.10	<0.10	<0.10	<0.10	0.2	<0.10	<0.10
S	(wt %)	0.61	0.03	0.05	0.21	0.81	1.15	0.02	0.03	0.49	BDL	0.02	0.26	0.78	0.03	0.62	0.36
Se	ppm	3.6	1.5	1.1	2.7	6.7	4.6	1.3	1.5	2.2	2	1.6	1.5	13.6	2.2	4.9	4.7
Ni	ppm	2710	230	750	1210	2580	2170	60	230	1990	BDL	290	60	4260	240	3780	30
Cu	ppm	1260	167	136	1050	1330	1910	9	35	1450	BDL	3	96	2920	BDL	2560	882
Mn	ppm	1020	1280	1380	1280	1790	1380	1850	1860	1490	972	1850	1500	3710	4070	2290	2190
Pd	ppb	1730	1	617	1550	2310	3640	2	32	2480	1	51	3	2890	4	5940	4090
Pt	ppb	1360	BDL	438	1120	1550	10200	4	17	1550	BDL	14	3	1440	3	3750	2370
Au	ppb	302	BDL	32	100	221	600	3	12	214	7	9	BDL	9	3	496	117
Ba	ppm	BDL	BDL	BDL	BDL	4	BDL	BDL	BDL	BDL	BDL	9	47	BDL	BDL	BDL	BDL
Th	ppm	0.1	0.2	0.3	0.3	0.4	1	2.9	3.5	0.4	20.2	1.4	1.3	0.2	BDL	0.2	9.7
U	ppm	BDL	BDL	BDL	BDL	BDL	BDL	BDL	BDL	BDL	11	0.2	0.2	0.5	BDL	BDL	1.5
Nb	ppm	BDL	BDL	BDL	BDL	BDL	BDL	BDL	2.5	BDL	4.8	BDL	BDL	BDL	BDL	BDL	BDL
Ta	ppm	BDL	BDL	BDL	BDL	BDL	BDL	BDL	BDL	BDL	0.2	BDL	BDL	BDL	BDL	BDL	0.4
Sr	ppm	8	12	12	23	6	10	25	24	9	BDL	13	14	22	8	12	9
Y	ppm	3.1	8.2	5.7	4.9	5.9	11.8	6.3	13.5	9.6	11.7	11.1	19.3	4.7	2.6	7.1	16.9
Cr	ppm	1330	330	60	BDL	1720	1170	BDL	500	60	BDL	410	70	160	BDL	BDL	BDL
Pb	ppm	3.9	4.4	5.2	14.9	4.9	3.9	1.4	2.2	3.5	35.1	3.3	10	2.2	BDL	7.7	5.1
La	ppm	1	2.3	1	2.2	2.1	4.7	1.9	5.3	1.9	3.6	6.9	3.8	3.1	2.8	2.8	4.4
Ce	ppm	2.3	6.7	3	5.3	4.7	11.9	4.9	13.4	4.9	7.2	14.1	10.8	8	5.4	6.7	10.2
Pr	ppm	0.3	1	0.4	0.6	0.6	1.7	0.6	1.7	0.7	0.8	1.8	1.6	1	0.6	0.9	1.3
Nd	ppm	1.2	4.7	1.8	2.6	2.4	7.2	2.5	7.6	3.5	3.1	8	7.5	4	2.1	3.7	5.3
Sm	ppm	BDL	1.2	0.2	0.2	0.4	1.8	0.4	1.7	0.7	0.8	1.8	2.1	0.7	0.2	0.9	1.4
Eu	ppm	0.1	0.4	0.1	0.1	0.2	0.3	0.2	0.4	0.3	0.3	0.8	0.6	0.1	0.2	0.3	0.3
Gd	ppm	0.5	1.6	0.8	0.9	0.9	2.3	1.1	2.6	1.5	1.6	2.3	3.1	1.1	0.5	1.3	2.1
Tb	ppm	BDL	0.3	0.2	0.2	0.1	0.3	0.3	0.5	0.3	0.3	0.3	0.6	0.2	BDL	0.2	0.4
Dy	ppm	0.5	1.6	1.2	1.1	1	2.2	1.5	2.7	1.9	1.9	2.1	3.5	1.1	0.4	1.3	2.9
Ho	ppm	BDL	0.4	0.2	BDL	0.2	0.5	0.3	0.6	0.4	0.4	0.5	0.8	0.2	BDL	0.3	0.7
Er	ppm	0.4	0.9	0.4	0.4	0.7	1.3	0.5	1.5	1	1.2	1.2	2.1	0.6	0.3	0.7	1.9
Tm	ppm	BDL	0.1	BDL	BDL	0.1	0.2	BDL	0.2	0.1	0.2	0.2	0.3	BDL	BDL	0.1	0.3
Yb	ppm	0.3	0.7	0.4	0.3	0.7	1.1	0.4	1.2	0.8	1.3	1	1.6	0.3	0.3	0.6	1.6
As	ppm	BDL	BDL	8	BDL	BDL	5	BDL	BDL	BDL	BDL	BDL	BDL	BDL	BDL	BDL	BDL
Bi	ppm	BDL	BDL	BDL	BDL	BDL	BDL	BDL	BDL	BDL	BDL	BDL	BDL	BDL	BDL	BDL	BDL
Te	ppm	BDL	BDL	BDL	BDL	BDL	BDL	BDL	BDL	BDL	BDL	BDL	BDL	BDL	BDL	BDL	BDL

BDL: below detection limit.

**Table 2.** Representative rim and core compositions of clinopyroxene measured by electron microprobe.

Sample ID	OY541-3b	OY541-3b	1 $\sigma$ * ( <i>n</i> = 55)	OY541-4b1	OY541-4b1	1 $\sigma$ * ( <i>n</i> = 51)	OY541-7c	OY541-7c	1 $\sigma$ * ( <i>n</i> = 33)
Spot#	105	109		50	75		44	33	
Location	rim	core		rim	core		rim	core	
SiO <sub>2</sub>	54.11	54.57	0.92	54.66	53.68	0.79	55.94	55.62	1.7
TiO <sub>2</sub>	0.05	0.08	0.15	0.02	0.15	0.14	0	0	0.16
Al <sub>2</sub> O <sub>3</sub>	0.75	0.47	0.88	0.76	1.45	0.6	0.14	0.06	2.29
FeO <sub>(tot)</sub>	2.11	0.98	4.08	1.74	2.97	2.28	0.18	0.09	0.61
MnO	0.12	0.15	0.08	0.19	0.24	0.06	0.96	0.64	1.19
MgO	17.08	17.59	3.32	16.92	16.18	1.01	18.03	18.32	1.25
CaO	24.82	24.62	7.81	24.51	24.43	1.38	24.49	24.86	0.65
NiO	0.05	0.05	0.03	0	0.03	0.02	0	0.01	0.01
Na <sub>2</sub> O	0.02	0	0.24	0.03	0.05	0.18	0.01	0.01	0.03
Total	99.11	98.51		98.83	99.18		99.75	99.61	
Structural formulae in atoms per formula unit (6 oxygens)									
Si	1.99	2.01		2.01	1.98		2.03	2.02	
Ti	0.00	0.00		0.00	0.00		0.00	0.00	
Al	0.03	0.02		0.03	0.06		0.01	0.00	
Fe <sup>2+</sup>	0.07	0.03		0.05	0.09		0.01	0.00	
Mn	0.00	0.01		0.01	0.01		0.03	0.02	
Mg	0.93	0.97		0.93	0.89		0.98	0.99	
Ca	0.98	0.97		0.97	0.96		0.95	0.97	
Ni	0.00	0.00		0.00	0.00		0.00	0.00	
Na	0.00	0.00		0.00	0.00		0.00	0.00	
Sum	4.00	4.00		4.00	4.00		4.00	4.00	

\* The standard deviation is based on both core and rim concentrations.

**Table 3.** Representative compositions of orthopyroxene relics in clinopyroxene measured by electron microprobe.

Sample ID	OY541-3b	OY541-3b	OY541-3b	OY541-7c
Spot#	129	138	145	146
SiO <sub>2</sub>	55.17	54.72	54.19	54.07
TiO <sub>2</sub>	0.05	0.21	0.09	0.18
Al <sub>2</sub> O <sub>3</sub>	1.11	1.13	1.55	1.43
FeO <sub>(tot)</sub>	15.14	15.18	15.69	15.50
MnO	0.33	0.34	0.36	0.34
MgO	26.82	26.68	26.71	26.77
CaO	1.65	1.61	0.74	1.16
NiO	0.07	0.09	0.06	0.11
Na <sub>2</sub> O	0.03	0.02	0.03	0.02
Total	100.4	99.98	99.42	99.58
Structural formulae in atoms per formula unit (6 oxygens)				
Si	1.98	1.98	1.97	1.96
Ti	0.00	0.01	0.00	0.01
Al	0.05	0.05	0.07	0.06
Fe <sup>2+</sup>	0.46	0.46	0.48	0.46
Mn	0.01	0.01	0.01	0.01
Mg	1.44	1.44	1.45	1.45
Ca	0.06	0.06	0.03	0.05
Ni	0.00	0.00	0.00	0.01
Na	0.00	0.00	0.00	0.00
Sum	4.00	4.00	4.00	4.00

**Table 4.** Sm-Nd isotope data of Platreef clinopyroxenites.

Sample ID	<sup>143</sup> Nd/ <sup>144</sup> Nd (Measured)	Error (2 s.d.)	Nd (ppm)	Sm (ppm)	<sup>147</sup> Sm/ <sup>144</sup> Nd	<sup>143</sup> Nd/ <sup>144</sup> Nd (initial) <sup>a</sup>	Error <sup>b</sup>	eNd (CHUR) <sup>c</sup>
OY541-1	0.511797	0.000004	1.198	0.313	0.1579	0.509655	0.000027	−6.16
OY541-3a	0.511513	0.000004	1.709	0.389	0.1375	0.509647	0.000024	−6.32
OY541-3b	0.511662	0.000004	2.271	0.567	0.1509	0.509615	0.000026	−6.94
OY541-4a	0.511638	0.000004	6.519	1.59	0.1474	0.509638	0.000026	−6.49
OY541-8	0.511759	0.000004	5.481	1.419	0.1564	0.509636	0.000027	−6.52

<sup>a</sup> Initial isotope data are calculated assuming an age of 2061 Ma; <sup>b</sup> The errors on initial <sup>143</sup>Nd/<sup>144</sup>Nd ratios are propagated from the uncertainty on the measured (present-day) <sup>143</sup>Nd/<sup>144</sup>Nd and <sup>147</sup>Sm/<sup>144</sup>Nd ratios; <sup>c</sup> Assuming CHUR values of 0.512638 and 0.1967 for <sup>143</sup>Nd/<sup>144</sup>Nd and <sup>147</sup>Sm/<sup>144</sup>Nd respectively.

Sm and Nd concentrations were measured on a separate dissolution of rock powder using a Thermo Scientific X-Series 2 quadrupole inductively coupled plasma mass spectrometer. Approximately 0.05 g of sample was accurately weighed into a Teflon beaker, and digested in 1 mL 15 M HNO<sub>3</sub> and 3 mL 12 M HF for 12 h in sealed beakers on a hotplate at 80 °C. After cooling, 0.2 mL of HClO<sub>4</sub> was added to the sample, and the solution evaporated to incipient dryness at 120 °C. 2 mL of 15 M HNO<sub>3</sub> was added to the sample, and evaporated to near dryness, and this step was repeated twice before increasing the hotplate temperature to 150 °C and fuming off excess HClO<sub>4</sub>. The sample was then redissolved in 4 mL 15 M HNO<sub>3</sub> and 4 mL H<sub>2</sub>O, 2 drops of 12 M HF were added, and the sealed beakers left on a hotplate at 80 °C for 12 h. The samples were then placed in an ultrasonic bath for 30 min, before heating at 80 °C for another 12 h. At this stage, all samples were completely in solution. The sample solutions were then quantitatively transferred to 250 mL HDPE bottles and diluted to 200 g with MQ water to obtain a final solution of 2% HNO<sub>3</sub> + 0.002 M HF with a sample dilution factor of about 4000 and total dissolved solids of 250 µg/mL. All reagents used were distilled in Teflon stills, and diluted with MQ 18.2 MΩ water. Sample solutions were introduced into the ICP-MS through a Cetac Aridus 2 desolvating nebulizer system in order to reduce molecular interferences. An ESI

SC-2 DX FAST autosampler was used to reduce washout times between samples. The instrument was tuned using a 5-ppb solution of Be, In and U; the typical sensitivity for  $^{238}\text{U}$  was  $2 \times 10^6$  counts per second for a sample uptake rate of 50  $\mu\text{L}/\text{min}$ . The Ce/CeO ratio was typically  $> 4500$ . The instrument was calibrated using multielement solutions covering the relevant concentration range. A mixed Be, Rh, In and Bi solution (30, 10, 10 and 5 ppb) was mixed with the sample online and these elements used as internal standards to correct for instrumental drift. Procedural blanks analyzed during this work were negligible for all elements measured. The uncertainty for Sm and Nd concentrations is typically 2–5%, and the uncertainty for the Sm/Nd ratio approximately 1% based on repeated analysis of standards. The initial Nd isotope ratios in Table 4 were calculated assuming an age of 2.061 Ga, using the  $^{147}\text{Sm}/^{144}\text{Nd}$  ratios for individual samples determined by quadrupole ICP-MS. Errors on the initial Nd isotope ratios in Table 4 are calculated by propagating errors on both  $^{143}\text{Nd}/^{144}\text{Nd}$  and  $^{147}\text{Sm}/^{144}\text{Nd}$  through the age correction. For more analytical details see [63].

Trace-element concentrations of BMS were conducted by LA-ICP-MS at the GeoZentrum Nordbayern (Erlangen, Germany) using an Agilent 7500c Q-ICP-MS equipped with an UP 193nm FX New Wave Research Excimer Laser. The concentrations of the PGE ( $^{99}\text{Ru}$ ,  $^{101}\text{Ru}$ ,  $^{103}\text{Rh}$ ,  $^{105}\text{Pd}$ ,  $^{108}\text{Pd}$ ,  $^{189}\text{Os}$ ,  $^{193}\text{Ir}$ ,  $^{195}\text{Pt}$ ) as well as  $^{197}\text{Au}$ ,  $^{107}\text{Ag}$ ,  $^{185}\text{Re}$ ,  $^{59}\text{Co}$ ,  $^{60}\text{Ni}$ ,  $^{61}\text{Ni}$ , and  $^{63}\text{Cu}$  in BMS were determined by in situ spot analyses on sulphides in three polished thin sections and five polished ore sections. He and Ar functioned as carrier gas with a flux of 0.9 l/min. The plasma power was 1100 W and Ar was used as plasma gas with a flux of 14.9 l/min. Measurements were undertaken with a repetition rate of 19 Hz for 25 s after a background scanning for 20 s. The laser energy settings are configured at 0.45 GW/cm<sup>2</sup> (fluence: 2.24 J/cm<sup>2</sup>). Beam diameters were varied between 15  $\mu\text{m}$  and 50  $\mu\text{m}$  depending on grain size. External standard materials include PO724 B2 SRM (Sulphide-standard Memorial University Newfoundland), (Fe,Ni)1-xS [64] and Mass1 (USGS) for BMS. S was analyzed by EMP and was used as internal standard for sulphide analysis. Reproducibility for SRM was  $< 9\%$  and the accuracy, tested by ablating the PGE SRM FeNiS standard, was  $< 15\%$ . The data were processed using GLITTER 4.4.4 software (Macquarie Research Ltd., Sydney, Australia, 2000). Detection limits for the analyses of sulphide minerals are given in Table 5. Various interferences caused by the carrier and plasma gas Ar lead to falsified results for Ru, Rh and Pd, if not corrected.  $^{99}\text{Ru}$  is interfered by  $^{63}\text{Cu}^{36}\text{Ar}$  and  $^{59}\text{Co}^{40}\text{Ar}$ ,  $^{101}\text{Ru}$  by  $^{63}\text{Cu}^{38}\text{Ar}$ ,  $^{65}\text{Cu}^{36}\text{Ar}$  and  $^{61}\text{Ni}^{40}\text{Ar}$ ,  $^{103}\text{Rh}$  by  $^{63}\text{Cu}^{40}\text{Ar}$ ,  $^{65}\text{Cu}^{38}\text{Ar}$  and  $^{67}\text{Zn}^{36}\text{Ar}$ ,  $^{105}\text{Pd}$  by  $^{65}\text{Cu}^{40}\text{Ar}$  and  $^{108}\text{Pd}$  by  $^{68}\text{Zn}^{40}\text{Ar}$ .  $^{108}\text{Pd}$  is additionally interfered by  $^{108}\text{Cd}$ . Due to the extremely low relative isotope abundance of  $^{36}\text{Ar}$  (0.337%) and  $^{38}\text{Ar}$  (0.063%), their effect on  $^{99}\text{Ru}$ ,  $^{101}\text{Ru}$  and  $^{103}\text{Rh}$  is negligible. Only a few counts per second for Zn were recorded for the sulphides in this study, thus excluding a significant influence. However, a remarkable impact occurs for CuAr on  $^{103}\text{Rh}$  and  $^{105}\text{Pd}$  when analyzing chalcopyrite. The argide-unaffected  $^{108}\text{Pd}$  for chalcopyrite was used, which nonetheless required a correction for the elemental interference from  $^{108}\text{Cd}$ . A PGE-free hydrothermal chalcopyrite was analyzed to check the  $^{63}\text{Cu}^{40}\text{Ar}$  and the  $^{65}\text{Cu}^{40}\text{Ar}$  production. Since the interference of  $^{61}\text{Ni}^{40}\text{Ar}$  lead to an overestimation of  $^{101}\text{Ru}$  in pentlandite,  $^{99}\text{Ru}$  was used instead to obtain the correct concentration. Numerous measurements of pentlandite yielded  $\sim 0.5$  ppm as the lowest concentration for Ru, indicating that this value solely originated from the Ni-argide. This concentration was always within the  $1\sigma$  error margin for every analysis.

In situ quadruple sulphur isotope ratios (V-CDT) were measured using a CAMECA IMS1280 large-geometry ion microprobe at the Center for Microscopy, Characterization and Analysis (CMCA), The University of Western Australia (Table 6). The ion microprobe was operated in multicollection mode using a Cs<sup>+</sup> primary beam with an intensity of  $\sim 3\text{--}4$  nA in Gaussian mode that interacted with the sample at 20 keV during the quadruple sulphur isotope analyses. Each analysis consists of a 25 four-second cycles acquisition. The analytical session was monitored in terms of drift using two bracketing standards every 6 sample analyses. Instrumental mass fractionation (IMF) was corrected using the matrix matched reference materials for pentlandite (VMSO), pyrrhotite (Alexo), and chalcopyrite (Nifty-b); reference values can be found in [65]. IMF correction follows the procedure described in [66].

**Table 5.** Minimum, maximum and median PGE concentrations (in ppm) in the BMS (from LA-ICP-MS).

Sample ID	n	Os			Ir			Ru			Rh			Pt			Pd		
Mineral		Min	Max	Median	Min	Max	Median	Min	Max	Median	Min	Max	Median	Min	Max	Median	Min	Max	Median
OY541-1		serpentinized clinopyroxenite																	
Pyrrhotite	39	BDL	1.43	0.65	BDL	2.37	0.81	BDL	7.10	2.00	BDL	BDL	BDL	BDL	0.76	0.56	BDL	0.93	0.93
Chalcopyrite	33	BDL	0.94	0.94	BDL	0.25	0.20	BDL	BDL	BDL	BDL	BDL	BDL	BDL	0.81	0.55	BDL	0.98	0.21
Pentlandite	30	BDL	3.80	1.37	BDL	3.10	1.86	BDL	27.7	6.76	BDL	16.9	4.39	BDL	2.72	0.65	27.4	566	252
OY541-3a		clinopyroxenite																	
Pyrrhotite	18	BDL	2.23	1.52	BDL	3.78	2.68	BDL	22.8	6.98	BDL	0.35	0.35	BDL	0.89	0.62	BDL	0.74	0.72
Chalcopyrite	43	BDL	3.77	1.14	BDL	7.48	1.05	BDL	27.6	3.28	BDL	BDL	BDL	BDL	7.30	0.68	BDL	21.21	0.29
Pentlandite (I)	10	BDL	0.55	0.55	0.75	1.80	0.91	2.98	4.69	3.41	8.55	9.91	9.12	BDL	1.48	1.48	74.7	85.6	83.0
Pentlandite (II)	21	0.42	21.7	3.11	1.60	12.2	4.71	0.80	98.7	16.8	0.35	55.0	16.6	0.42	5.16	1.21	162	198	175
OY541-3b		serpentinized clinopyroxenite																	
Pyrrhotite	36	BDL	3.74	1.09	BDL	4.72	1.96	BDL	38.6	4.87	BDL	0.08	0.08	BDL	1.11	0.72	BDL	0.27	0.15
Chalcopyrite	18	BDL	BDL	BDL	BDL	0.57	0.40	BDL	0.66	0.66	BDL	BDL	BDL	BDL	0.29	0.29	0.05	2.65	0.13
Pentlandite (I)	11	1.84	1.91	1.19	1.84	2.72	2.24	7.45	12.8	8.38	9.37	24.5	11.9	BDL	0.60	0.52	192	227	214
Pentlandite (II)	8	BDL	0.46	0.46	BDL	BDL	BDL	0.90	2.81	1.23	1.37	3.62	1.94	BDL	0.49	0.49	496	631	538
Pentlandite (III)	28	BDL	3.41	1.83	0.65	7.55	3.37	0.42	23.4	12.3	4.53	62.6	14.1	BDL	BDL	BDL	100	169	134
OY541-4a		serpentinized clinopyroxenite																	
Pyrrhotite	58	BDL	1.49	0.90	BDL	3.28	1.18	BDL	20.75	5.60	BDL	2.31	0.19	BDL	2.00	1.00	BDL	3.60	0.12
Chalcopyrite	9	BDL	BDL	BDL	BDL	BDL	BDL	BDL	BDL	BDL	BDL	BDL	BDL	BDL	BDL	BDL	BDL	BDL	BDL
Pentlandite	39	BDL	2.95	1.24	BDL	26.1	2.65	BDL	34.2	11.7	BDL	65.4	18.8	BDL	0.16	0.16	125	341	246
OY541-8b		serpentinized clinopyroxenite																	
Chalcopyrite	2	BDL	BDL	BDL	BDL	BDL	BDL	BDL	BDL	BDL	BDL	BDL	BDL	BDL	BDL	BDL	BDL	BDL	BDL
Pentlandite	47	BDL	8.28	1.26	BDL	12.5	0.80	0.73	41.5	9.98	0.13	67.5	1.26	BDL	5.40	0.84	89.2	185	112

BDL: below detection limit.

**Table 6.** Median values and standard deviations of sulphur isotope compositions.

Title		$\delta^{33}\text{S}$ (‰, V-CDT)		$\delta^{34}\text{S}$ (‰, V-CDT)		$\Delta^{33}\text{S}$		
Sample	Mineral	Median	1 $\sigma$	Median	1 $\sigma$	Median	1 $\sigma$	n
OY541-1	Ccp	2.19	0.05	3.78	0.12	0.25	0.03	9
	Po	1.70	0.12	2.80	0.21	0.26	0.02	4
	Pn	1.51	0.03	2.42	0.07	0.28	0.02	4
OY541-3b	Ccp	1.68	0.09	2.95	0.17	0.16	0.04	8
	Po	1.55	0.16	2.52	0.34	0.27	0.04	16
	Pn	1.04	0.11	1.66	0.22	0.19	0.06	19
OY541-4a	Ccp	1.68	0.02	2.96	0.03	0.16	0.02	4
	Po	1.56	0.11	2.60	0.20	0.21	0.04	21
	Pn	1.03	0.37	1.60	0.64	0.23	0.06	11
OY541-8b	Ccp	2.25	0.06	3.81	0.13	0.30	0.04	3
	Pn	1.81	0.52	2.98	0.95	0.24	0.10	8

## 4. Results

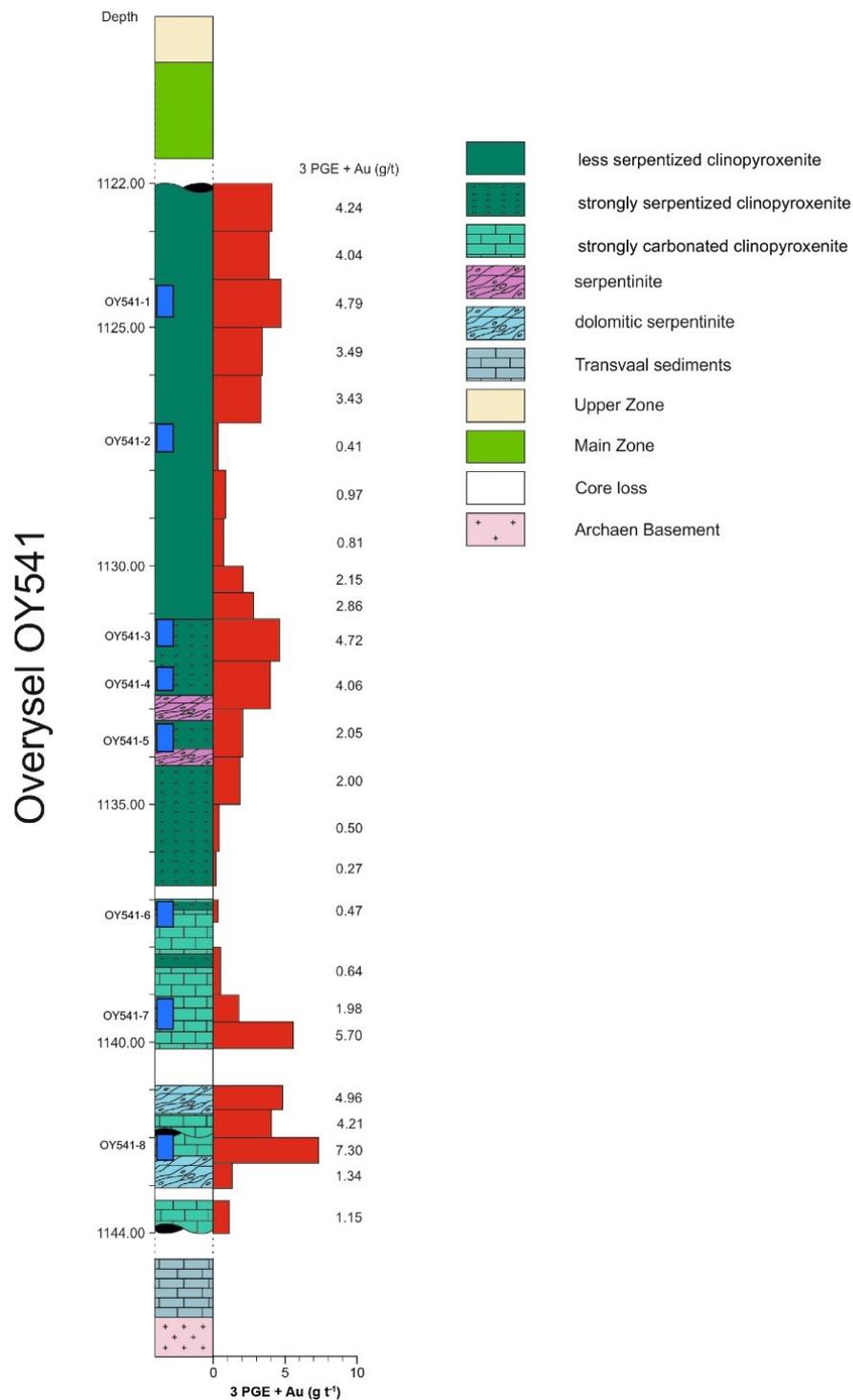
### 4.1. Petrography and Mineral Chemistry

The drill core OY541 at Overysel comprises mineralized, serpentinized and carbonated clinopyroxenites (Figure 2). The least altered granoblastic clinopyroxenite (samples OY541-2, -3a) contains 90–95 vol % idioblastic diopside (Figure 3A; Table 2). Clinopyroxene of the uppermost part of the sampled succession contains enstatite relic inclusions (Figure 3B; Table 3). The compositions of clinopyroxene are chemically not perfectly homogeneous, but do not reveal any consistent pattern from core to rim (Table 2). The rim and core compositions, and the standard deviations, for three samples are given in Table 2 and Figure 4. Diopside from sample 3b additionally contains exsolution lamellae of augite and/or pigeonite (Figure 4). Some clinopyroxenite samples also display slight ductile and/or brittle deformation overprints, as is reflected by kinked clinopyroxene bands and subsequent crosscutting fractures filled with serpentine and carbonate. Pyroxene grains display varying degrees of patchy serpentinization and chloritization, which are in places associated with actinolite. Accessory phases comprise biotite, relic olivine as well as secondary calcite and dolomite usually occurs with serpentine and talc along crosscutting veinlets.

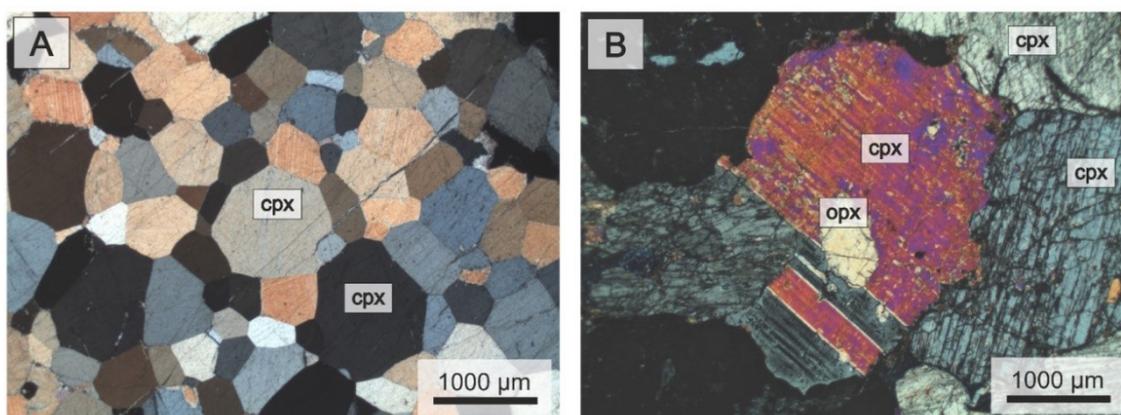
Highly altered clinopyroxenite (samples OY541-1, -3b, -4a, b, -5, -6, -7a, b, c, -8), which is often intimately interlayered with less-altered clinopyroxenite, is characterized by pervasive serpentinization, chloritization and/or carbonation, accompanied by serpentine, talc, chlorite and carbonate veining. In some places, clinopyroxene is replaced by tremolite fibroblasts. In the most pervasively altered samples clinopyroxene only occurs as relic phase.

### 4.2. Ore Petrography

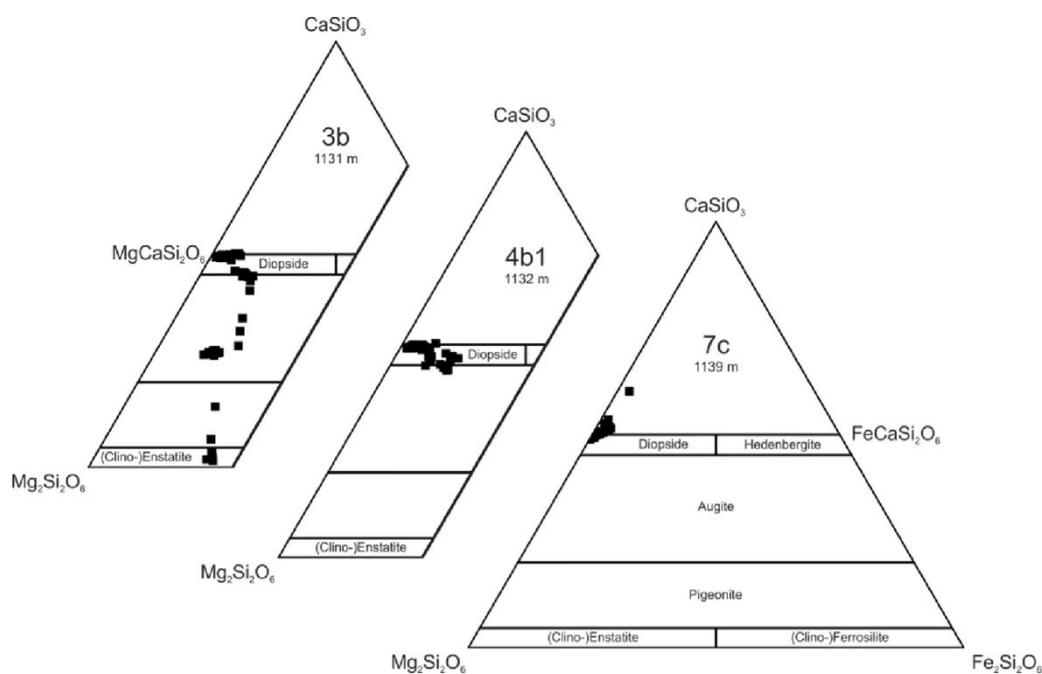
According to the terminology of Barnes, Mungall, Le Vaillant, Godel, Leshner, Holwell, Lightfoot, Krivolutskaya and Wei [67], sulphide mineralization within the footwall clinopyroxenites is heterogeneous in its distribution, and occurs as “disseminated” (up to about 5 vol % sulphides), poly-phased interstitial millimetre- to centimetre-sized grains, which locally grade into patchy “net-textured” ore (between 30 and 70 vol % sulphides), displaying sulphide-rich and sulphide-poor domains on the cm-scale. The main ore minerals are pentlandite, pyrrhotite and chalcopyrite, with accessory (titano) magnetite, cubanite and mackinawite. Pyrrhotite shows rare pentlandite exsolution lamellae, while chalcopyrite shows cubanite exsolution in places. Interstitial blebs and accumulations of the disseminated and patchy net-textured sulphides usually range from 200 to 600  $\mu\text{m}$ , but may locally reach several centimetres in size (Figure 5A). Almost all BMS throughout the drill core are sub-to anhedral, with serrated grain boundaries, indicating post-solidification dissolution and replacement by secondary silicates and magnetite (Figure 5 A,B).



**Figure 2.** Simplified stratigraphy of the borehole (OY541) showing the clinopyroxenite package, which forms the footwall to the Platreef at Overysel based on the present study and core log information provided by Anglo Platinum. Grade and distribution of mineralization is indicated by red boxes (3PGE + Au). The numbers attached on the left give the sample depth in meters. The blue rectangles mark the sections studied here.

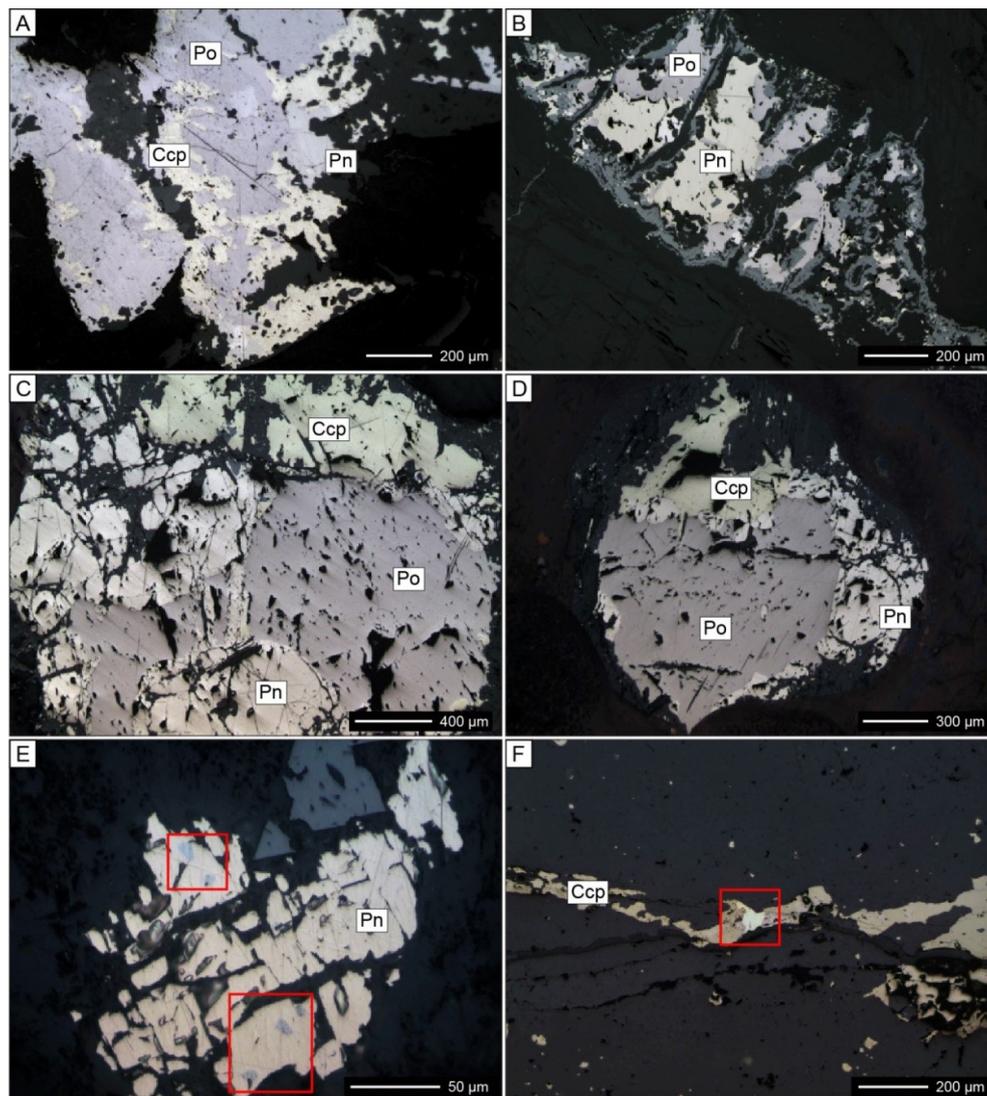


**Figure 3.** (A) Euhedral recrystallized clinopyroxenes displaying a granoblastic texture 1; (B) Twinned diopside with an enstatite inclusion (mineral abbreviations follow [68]).



**Figure 4.** Calculated pyroxene endmembers (from EMP analysis). The analyses that plot between enstatite and diopside in samples 3B (top left) are across exsolution lamellae.

In addition, rare globular sulphides, which to our knowledge have not been reported from the Platreef sulphide ore before, occur as rounded, interstitial aggregates, which are 100 to ca. 1000  $\mu\text{m}$  in diameter (Figure 5C,D). The differentiated globules display chalcopyrite in its upper part, separated by a smooth meniscus from pentlandite and pyrrhotite occupying the lower portion of the sulphide blebs. The globules may or may not have silicate caps, which comprise variable proportions of clinopyroxene, serpentine and magnetite, above the sulphide aggregates (Figure 5C,D). Moreover, the BMS infrequently show a crimped angular habit due to the recrystallization of the surrounding clinopyroxene. The BMS assemblage of the deeper core section is dominated by pentlandite, while pyrrhotite is rather rare. The pentlandite in the strongly altered samples is partly replaced by secondary alteration phases, such as serpentine and carbonates, and also occurs embedded along the cleavage planes of clinopyroxene, and along veins associated with secondary serpentine and carbonates (Figure 5E). Veinlet-controlled mineralization is dominated by chalcopyrite, and is usually associated with carbonate minerals, magnetite or secondary serpentine (Figure 5F). Thus, these textures reflect a complex multistage mineralization history.



**Figure 5.** Sulphide textures. (A) Disseminated sulphides with serrated grain boundaries; (B) Sulphide relics displaying replacement along the rim and fractures by magnetite and serpentinite indicating post-solidification replacement. (C) Differentiated globular sulphide characteristically showing pyrrhotite and pentlandite at the base separated by a smooth meniscus from chalcopyrite at the top of the bleb. (D) Clinopyroxene-capped differentiated sulphide globule with a flat base. (E) Pentlandite relics within a serpentine matrix displaying inclusions of Pd-bismuthotellurides (within red boxes). (F) Chalcopyrite bearing veinlet with gold grain (Mineral abbreviations follow [68]).

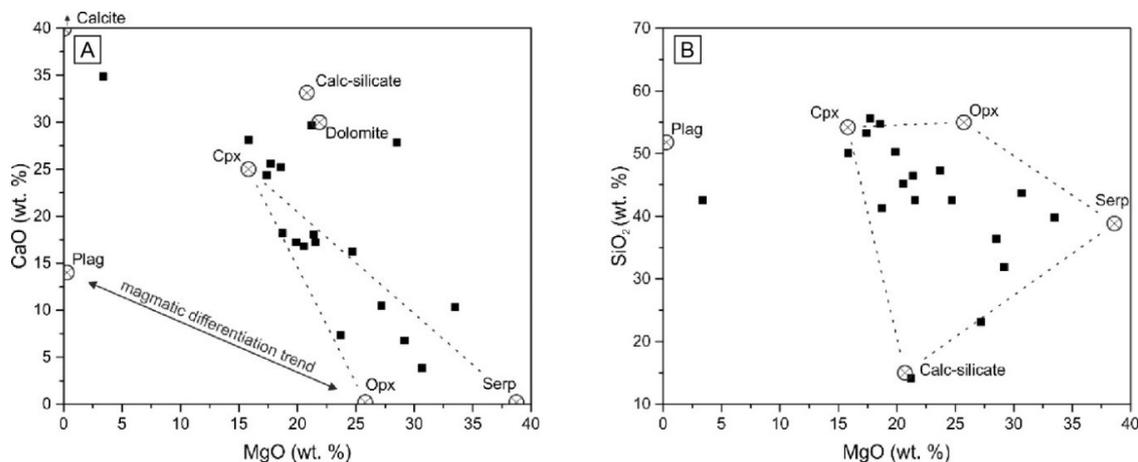
The identified PGM were bismuthotellurides, sperrylite and stibiopalladinite, which were found to be almost always associated or enclosed within the BMS blebs (Figure 5E) and, in addition, one grain of gold was identified on a late fracture filled with chalcopyrite (Figure 5F).

#### 4.3. Bulk-Rock Chemistry

The major- and trace-element compositions of the Platreef clinopyroxenite were described in detail by Harris and Chaumba [3]; Armitage, McDonald, Edwards and Manby [15]; McDonald, Holwell and Armitage [62]; Holwell, McDonald and Armitage [20]; and Kinnaird, Hutchinson, Schurmann, Nex and de Lange [22], and thus only the characteristics relevant for this study are discussed here.

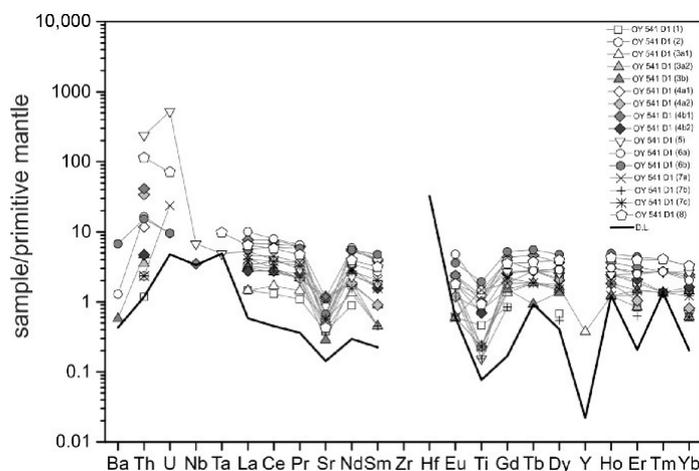
Binary variation diagrams of CaO vs. MgO, and SiO<sub>2</sub> vs. MgO (Figure 6), are consistent with the clinopyroxene-dominated mineralogy and distinct from the “magmatic trend”, representing magmatic variations due to varying plagioclase and orthopyroxene proportions, commonly found

in the pyroxenitic units (Figure 6A) [6,48]. The clinopyroxenite samples investigated in this study plot off the “magmatic trend” and define a steeper slope in the CaO vs. MgO diagram (Figure 6A). Furthermore, the low Cr (<1700 ppm; Table 1) contents are in accordance with a non-magmatic origin of the clinopyroxenites [3] (see also Figure 4 in [6]). This is further supported by their lower SiO<sub>2</sub>, FeO, Na<sub>2</sub>O and Cr, but higher CaO concentrations, than those of primary magmatic Platreef pyroxenites and norite (see [6]).



**Figure 6.** Binary variation diagrams of (A) CaO vs. MgO; (B) SiO<sub>2</sub> vs. MgO (black squares represent the bulk chemistry of the samples studied here). Fields framed by stippled lines display the range of clinopyroxenite composition from Sandsloot (modified after [3]). The plagioclase, dolomite, calcite, serpentine and calcsilicate compositions are compiled from Pronost, Harris and Pin [6]; McDonald and Holwell [28]; and Deer, Howie and Zussman [69]. The “magmatic trend” is a correlation between CaO and MgO, displaying a magmatic variation due to varying plagioclase and orthopyroxene proportions cf. [3,6].

Multi-element diagrams of the clinopyroxenite samples display a slight negative slope, with negative anomalies of Ba, Nb, Ta, Sr, Ti and Y, while an Eu anomaly is absent (Figure 7). All investigated clinopyroxenite samples display a similar REE pattern, involving a slight enrichment of light REE (LREE) relative to heavy REE (HREE) (La<sub>N</sub>/Yb<sub>N</sub> ranging from 1.6 to 7.0; Dy<sub>N</sub>/Yb<sub>N</sub> ranging from 0.9 to 2.4). These patterns are similar to those of other clinopyroxenite and norite samples from Overysel and Sandsloot, as reported by Harris and Chaumba [3], Pronost, Harris and Pin [6] and McDonald and Holwell [28].



**Figure 7.** Multi-element diagram of trace elements (median values from LA-ICP-MS) of clinopyroxenite samples normalized against primitive mantle after McDonough and Sun [70]. D.L. = detection limit.

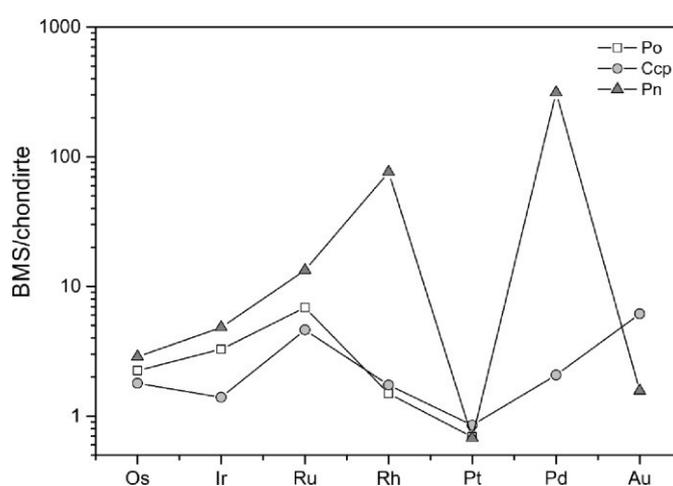


**Table 7.** Median values for S, Ni, and Cu (EMP), and Pd and Pt (LA-ICP-MS) in Pn, Po and Ccp, as a basis for the mass-balance calculation.

Sample ID	S Pn (%)	S Po (%)	S Ccp (%)	Ni Pn (%)	Cu Ccp (%)	Pd in Pn (ppm $\pm 1\sigma$ )	Pd in Po (ppm)	Pd in Ccp (ppm)	Pt in Pn (ppm)	Pt in Po (ppm)	Pt in Ccp (ppm)
OY541-1	33.3	37.3	34.6	30.4	34.7	252.34 $\pm 131$	b. d. l.	0.21	0.64	0.56	0.55
OY541-3a	33.3	38.4	34.7	32	35	167.99 $\pm 55$	0.72	0.29	1.25	0.62	0.68
OY541-3b	33.3	38.3	34.9	31.5	35	308.88 $\pm 158$	0.15	0.13	0.5	0.72	0.29
OY541-4a	33.3	38.6	34.6	32	35	246.00 $\pm 54$	0.12	b. d. l.	0.164	1	b. d. l.
OY541-8b	33.2	n. c.	34.6	36.8	35.2	111.87 $\pm 14$	n. p.	b. d. l.	0.84	n. p.	b. d. l.

n. p.: not present; b. d. l.: below the detection limit.

Pentlandite displays the highest median PGE concentrations of all BMS in the clinopyroxenite (Table 4), with median values of 218 ppm Pd, 9.9 ppm Rh, 9.5 ppm Ru, 1.4 ppm Os, 2.2 ppm Ir and 0.7 ppm Pt. Furthermore, pentlandite commonly exhibits high Ru, Rh and Pd, and low Pt contents (Figure 9), whereas chalcopyrite and pyrrhotite display only slightly elevated Ru (~4 ppm) contents and low Os, Ir, Rh, Pt and Pd contents, similar to previous datasets (cf. [9,12,21]). The PGE patterns of the pentlandite from Overysel are similar to those from Sandsloot, and are characterized by Pt concentrations < 1 ppm, and concentrations of Ru and Rh ranging from 28 ppm to 19 ppm, respectively (cf. [9,12]). In contrast, pentlandite PGE concentrations from Turfspruit are commonly below the detection limit, except for Pd, which often has concentrations of ~10 ppm [22]. The pentlandite PGE concentrations from Aurora display values below tens of ppm for Pt, Rh, Ru, Os and Ir, and 30 ppm for Pd [29], whereas Pd concentrations in pentlandite for the Flatreef can range up to 500 ppm [19]. The pentlandite Pd concentrations display a broad range concerning the location (i.e., Overysel, Sandsloot and Aurora) and the host rock type (cf. [9,12]). The clinopyroxenite and calcsilicate rock samples from Sandsloot show high average values of ca. 360 ppm, whereas the clinopyroxenite at Overysel displays lower average values of ca. 218 ppm (this study) and 120 ppm [12].

**Figure 9.** Chondrite normalized median PGE concentrations (from LAICP-MS) hosted by BMS. Normalization values were taken from McDonough and Sun [70].

4.7. Mass-Balance Calculation

The relative quantities of BMS-hosted PGE in individual bulk samples can be constrained using a mass-balance based on BMS and whole-rock Pd, Pt, S, Cu and Ni concentrations (Table 7) [8,12,74]. The weight fraction (in %) of the PGE in BMS ( $P_{sul}^i$ ) is calculated using:

$$P_{sul}^i = \left( F_{sul} \times \frac{C_{sul}^i}{C_{wr}^i} \right) \times 100 \tag{1}$$

where  $F_{sul}$  denotes the BMS weight fraction per sample,  $C_{sul}$  and  $C_{wr}$  are the median concentrations of the element  $i$  in the BMS and whole rock, respectively. The weight fractions of pentlandite, chalcopyrite and pyrrhotite are calculated based on the assumption that Ni is exclusively hosted by pentlandite, Cu is exclusively hosted by chalcopyrite, and that residual S represents the pyrrhotite content [8]:

$$F_{Pn} = (Ni_{wr}/Ni_{Pn}) \tag{2}$$

$$F_{Ccp} = (Cu_{wr}/Cu_{Ccp}) \tag{3}$$

$$F_{Po} = (S_{wr} - (S_{Ccp} \times F_{Ccp}) - (S_{Pn} \times F_{Pn}))/S_{Po} \tag{4}$$

Cubanite and mackinawite were excluded from the calculation due to their low modal abundance. The whole-rock Ni content was adjusted to account for the Ni hosted in clinopyroxene. Further, mass-balance results are still error-prone, considering their dependency on (1) the assumption that all Ni is as pentlandite and clinopyroxene, neglecting other Ni-bearing phases such as serpentine, relic olivine, pyrrhotite, Ni-alloys and millerite, (2) uncertainties of the analytical chemistry, (3) the nugget effect and (4) a calculation based on median values for the PGE concentrations, which show clustering.

The calculated sulphide weight fractions and the petrography of five representative clinopyroxenite samples revealed a relatively constant pentlandite content throughout the drill core samples (Figure 10, Table 8). Chalcopyrite contents show minor variations, while pyrrhotite contents distinctly vary down-hole.

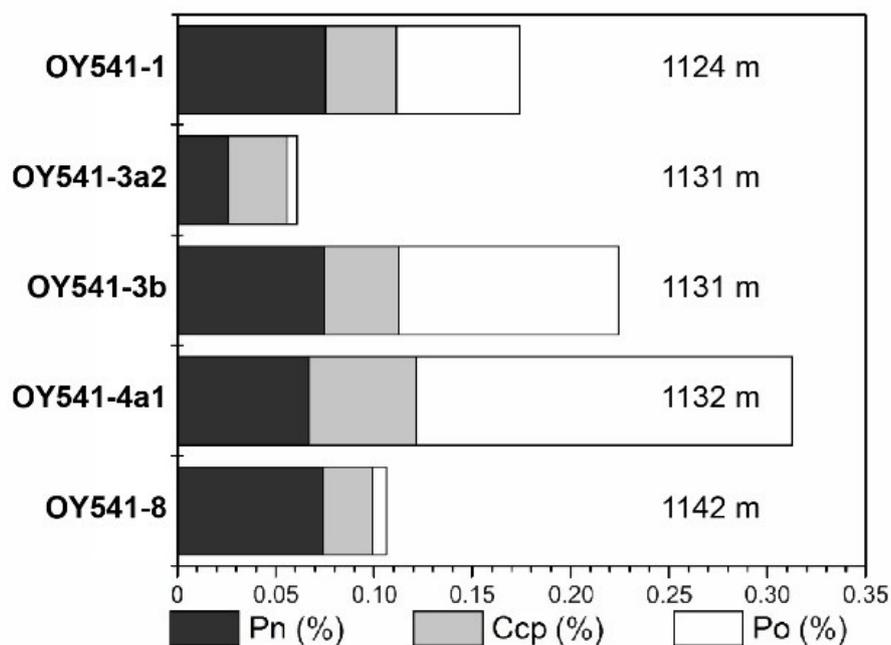


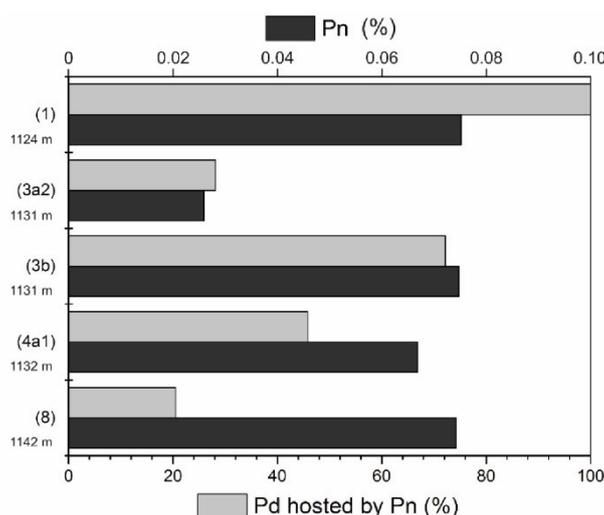
Figure 10. Calculated change of the base-metal sulphide content down-hole (data from Table 8).

**Table 8.** Calculated proportions of pentlandite, pyrrhotite, chalcopyrite and Pt, and Pd proportions (in %) hosted by pentlandite in Overysel clinopyroxenite samples.

Sample ID	Pn (%)	Po (%)	Ccp (%)	Population 1	Population 1	Population 2	Population 2	Population 3	Population 3	Median
				Pd (%)	Pt (%)	Pd (%)	Pt (%)	Pd (%)	Pt (%)	Pd (%)
OY541-1	0.075	0.062	0.038	106	0.91	n. c.	n. c.	n. c.	n. c.	106
OY541-3a	0.025	0.005	0.029	13.7	0.34	30.2	0.28	n. c.	n. c.	28.1
OY541-3b	0.074	0.111	0.038	63.2	0.23	166	0.21	39.2	n. c.	72.2
OY541-4a	0.067	0.19	0.054	45.8	0.01	n. c.	n. c.	n. c.	n. c.	45.8
OY541-8b	0.074	0.006	0.025	20.5	0.26	n. c.	n. c.	n. c.	n. c.	20.5

n. c.: not calculated.

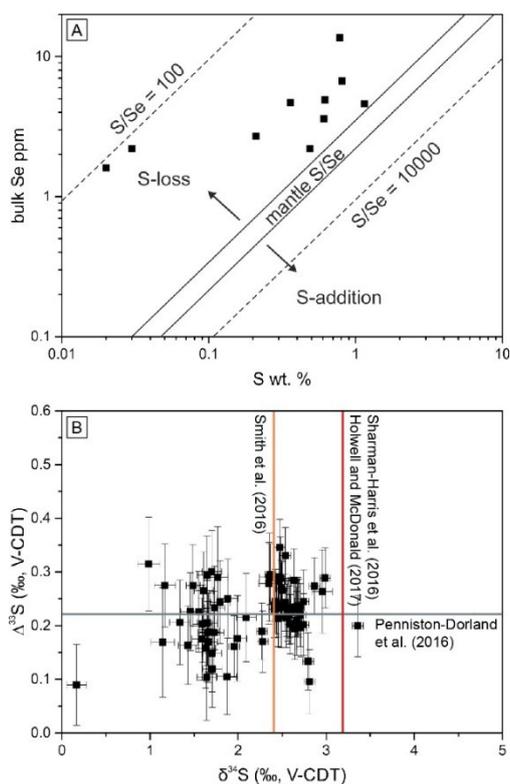
The calculated proportions (%) of Pt and Pd hosted by pentlandite are given in Table 8. Due to the clustering of the Pd contents in individual pentlandite grains, different populations were calculated to account for the possible range of the PGE distribution. Furthermore, the clustering of the Pd concentrations also results in high standard deviations for each population (Table 8). The mass-balance indicates that the pentlandite hosts variable fractions of the bulk-rock Pd—between ~110% (sample OY 541-1) and ~21% (OY 541-8a) throughout the drill core (Figure 11). In addition, chalcopyrite and pyrrhotite host only minor Pd contents, with maximum median values of 0.72 ppm each (Table 7).

**Figure 11.** Calculated Pd proportion hosted by calculated pentlandite (data from Table 8).

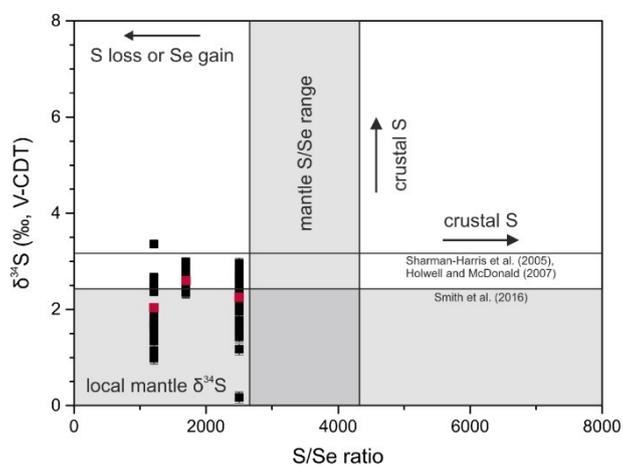
#### 4.8. S-Isotopes

Sulphur isotope ratios (Table 6) were measured in situ in chalcopyrite, pyrrhotite and pentlandite from four representative clinopyroxenite samples (OY 541-1, -3b, -4a and -8). However, in sample OY 541-8 sulphur isotope ratios (Table 6) were measured in situ only in chalcopyrite and pentlandite, since pyrrhotite is absent in this sample. Chalcopyrite consistently shows higher  $\delta^{34}\text{S}$  isotopic ratios (+3.81‰ to +2.95‰) than pyrrhotite and pentlandite. However, the relative difference in the  $\delta^{34}\text{S}$  (0.7‰ to 1.0‰  $\pm$  0.5‰) between pyrrhotite and pentlandite is consistent with that inferred for equilibrium S-isotope fractionation [65].

The analyzed pentlandite and pyrrhotite show  $\delta^{34}\text{S}$  signatures within or close to the pristine magmatic range of  $-1.8\text{‰}$  to  $+3.2\text{‰}$  [16,75–77]. The  $\Delta^{33}\text{S}$  range, from  $+0.11\text{‰}$  to  $+0.38\text{‰}$ , may indicate the assimilation of sedimentary sulphur in an early stage chamber prior to emplacement (Figures 12B and 13) [78].



**Figure 12.** (A) Sulphur wt % vs. bulk Se (ppm) from samples with S contents associated with substantial amounts of mineralization; (B) Sulphur isotope compositions of measured pyrrhotite and pentlandite. The grey line represents the  $\Delta^{33}\text{S}$  range of the pristine magma, whereas the orange and red lines represent the  $\delta^{34}\text{S}$  range of the pristine magma. Error bars are  $1\sigma$  [16,75,77,78].



**Figure 13.**  $\delta^{34}\text{S}$  and bulk rock S/Se systematics of sample OY541-1, OY541-3b and OY541-4a. Red squares represent median values of the samples. Horizontal lines represent different suggested  $\delta^{34}\text{S}$  ranges for a pristine Bushveld magma.

## 5. Discussion

### 5.1. Origin of the Clinopyroxenitic Units

Formation models of the Platreef clinopyroxenite vary, and current interpretations include magmatic and metamorphic processes, skarn-type contact metasomatism of the footwall dolostone, and formation of “hybrid-lithologies” by assimilation of sedimentary rocks in the magma. The described clinopyroxenite differs in modal compositions, and comprise strongly Ca-enriched units, which is

reflected by the presence of a calcsilicate assemblage consisting of varying amounts of diopside, wollastonite, olivine, garnet (andradite), calcite, amphiboles and spinel [5,6,20,22,36,62,79]. The distinct mineral assemblage—the common occurrence of dolostone xenoliths in the clinopyroxenite—and its contact with footwall Malmani Dolomite at Overysel and Sandsloot suggests that magma–carbonate interaction has played an important role in clinopyroxenite formation (e.g., [3,6,80]). Additionally, the clinopyroxenite is often pervasively carbonated and serpentinized, indicating post-solidification alteration by (hydrothermal) fluids [5,6,20,22,62,79].

The here-studied clinopyroxenite samples from hole OY541 contain a less diverse mineral assemblage than that documented at Sandsloot and Tweefontein (cf. [3,15,20,36,81]), and mainly comprise clinopyroxene, orthopyroxene and serpentine, and additional carbonates in the deeper part of the drill core. The occurrence of orthopyroxene inclusions and pigeonite exsolution lamellae in diopside supports a metasomatic formation of clinopyroxene at the expense of magmatic orthopyroxene, consistent with the observation at Sandsloot [20,82]. In addition, preliminary oxygen- and carbon-isotope ratios that were determined for the dolomite grains separated from two pervasively altered clinopyroxenite samples (OY541-7b and 7c), revealing  $\delta^{13}\text{C}$  values of  $-4.96\text{‰}$  and  $-5.88\text{‰}$ , and  $\delta^{18}\text{O}$  values of  $20.35\text{‰}$  and  $21.88\text{‰}$ , respectively (unpublished data). This stable isotope data is consistent with late hydrothermal alteration of these Platreef clinopyroxenite samples (cf. [3,6]).

The clinopyroxenite paragenesis is in accordance with the whole-rock chemical composition. The clinopyroxenite samples studied here, like those from Sandsloot, plot off the “magmatic trend” (Figure 6A) due to higher CaO/MgO, and lower SiO<sub>2</sub>, FeO and Na<sub>2</sub>O contents, compared to the typical magmatic Platreef pyroxenite and norite from Sandsloot, further supporting the non-magmatic origin of the clinopyroxenite (cf. [3,6,28]). This is also consistent with an overall relatively low Cr concentration (<1700 ppm). Moreover, clinopyroxenite  $\epsilon\text{Nd}(2.06\text{ Ga})$  values between  $-6.16$  and  $-6.94$  are similar to those of the magmatic pyroxenite and norite of the Platreef at Sandsloot and Overysel (average  $\epsilon\text{Nd} = -6.8 \pm 1.25$ ; [6]), and also the Main Zone [average  $\epsilon\text{Nd}(2.06\text{ Ga}) = -6.76 \pm 0.47$ ; [71]], indicating that clinopyroxenite formation at Overysel is unlikely to be related to the emplacement of a compositionally distinct melt. The clinopyroxenite was previously interpreted as indicating the assimilation of country rock, such as Malmani Dolomite (cf. [3,6,28]). However, dolostone assimilation and Ca-enrichment during ongoing pyroxene crystallization would have promoted the co-crystallization of orthopyroxene and clinopyroxene, and pyroxene–pyroxene exsolution, upon cooling, instead of the formation of clinopyroxene containing orthopyroxene relics (cf. [54]). Thus, these results rather support a post-magmatic metasomatic formation of clinopyroxenite, at the expense of former cumulate orthopyroxene.

## 5.2. Sulphide Mineralization and Distribution of PGEs

Even though there is broad agreement on the magmatic origin of the Platreef, its sulphide mineralization mechanism is still at the centre of the scientific debate [1,2,19,44]. In particular, the composition of the Platreef footwall’s clinopyroxenite (and thus the associated sulphide mineralization) fails to reconcile with a purely magmatic evolution, and suggests modification by country rock assimilation and/or interaction with hydrothermal fluids. Crustal contamination may have modified (cf. [3,6,56]), or even triggered, mineralization by increasing the S-saturation of the magma, via the assimilation of S-rich country rocks [36,75,83], while fluid infiltration may have redistributed mineralization (e.g., [20]).

We contribute to this ongoing discussion by presenting sulphide and silicate textural observations in combination with novel geochemical data. The described globular sulphides display a characteristic distribution of chalcopyrite, pyrrhotite and pentlandite in relation to the consistently flat base of the globules (Figure 4C,D). These observations are consistent with recrystallization from the former monosulphide solid solution, which crystallized from a sulphide liquid during downward transport of the blebs (e.g., [38,84]). The silicate caps that overlie the sulphide globules comprise variable proportions of clinopyroxene, and secondary magnetite and serpentine.

LA-ICP-MS results reveal pentlandite as the major carrier of Pd (Table 5; Figure 9), whereas chalcopyrite and pyrrhotite host almost no PGE [8,11,12,16,21,29,85]. High Pt and IPGE contents were interpreted to be due to discrete micro-inclusions in pentlandite, or due to the substitution of PGM for Ni and/or Fe in BMS (cf. Figure 4E) [11,12,21]. In case of the detection of micro-inclusions of PGM, individual analyses were removed from the median values. However, the PGE substitution for Ni and/or Fe is heterogeneous at the scale of a single sample (Table 8), but homogeneous at the single-grain scale. The variability of PGE concentrations may be related to variations in the S content at S-saturation, resulting in relatively diluted PGE concentrations during the abundant formation of BMS at high sulphur saturation (cf. [38]) (Figure 10). Therefore, the variations of pentlandite grain PGE concentrations on a thin section scale could be due to the presence of different generations of pentlandite. We cannot rule out that this variation in pentlandite PGE contents reflects a complex multistage mineralization, even though the chemically defined populations do not correlate with observed distinct pentlandite textures. Detailed 3D textural analysis by X-ray-computed tomography would be necessary to gain unambiguous textural evidence (cf. [67]). Furthermore, as indicated by the sulphide reaction-textures (i.e., their replacement by the late secondary carbonates and serpentine, as well as dissolution along the grain boundaries), post-solidification mobilization and redistribution may have contributed to the generation of the different pentlandite grain PGE populations.

S/Se ratios and S isotopes often provide important constraints for the source of S and the role of crustal contamination in triggering sulphide saturation in magmatic Ni–Cu–PGE sulphide deposits (Figures 12B and 13). Most Ni–Cu–PGE deposits show S/Se within or close to the mantle range of 2859 to 4350, whereas S/Se in crustal rocks is typically above 3500 [36,86–88]. Recent investigations on the Aurora limb show a wide range of BMS-S/Se ratios between 774 and 23384 (cf. [29]). However, the here-investigated bulk-samples show S/Se between 454 and 2500, i.e., lower than typical values of mantle and crustal rocks (Figures 12A and 13). Thus, S/Se ratios provide no evidence for a significant effect of crustal contamination on the mineralization, but contamination by sedimentary country rocks (for which we have no evidence in the investigated drill core) may be masked by syn- or post-magmatic processes. However, the BMS reaction-textures, and their frequent association with hydrothermal alteration products in the studied clinopyroxenite, suggest that the low S/Se results from secondary S-loss (cf. [29,88,89]). This is supported by a S/Se and S isotope study of primary and secondary sulphides from the Grasvalley Norite–Pyroxenite–Anorthosite (GNPA) member of the northern Bushveld limb [72,76]. These data revealed minor to no assimilation of crustal rock during the emplacement (based on  $\delta^{34}\text{S}$  values), and S-loss due to secondary processes, such as hydrothermal alteration, metamorphism and serpentinization [76,90,91]. Additional constraints on the sulphur origin may be derived from sulphur isotope ratios. The isotope ratios of the three analyzed sulphide phases in OY541 (chalcopyrite, pentlandite, pyrrhotite) are indicative of the mass-independent fraction of sulphur (MIFS), expressed as  $\Delta^{33}\text{S}$  (Figure 12B). MIFS is almost exclusively found in rocks older than the great oxidation event, i.e., older than 2.45 Ga [19,92–94]. This implies that the sulphide phases crystallized from pristine Bushveld magmas (i.e., 2.054 Ga) are not expected to show MIFS, unless sulphur was assimilated from a significantly older source.

## 6. Conclusions

The major- and trace-element characteristics of this clinopyroxenite, in conjunction with higher CaO/MgO ratios and lower SiO<sub>2</sub>, FeO and Na<sub>2</sub>O contents in comparison to those of typical igneous Platreef pyroxenites and norite from elsewhere, are not in accordance with a magmatic trend, and rather appear to be the result of a metasomatic origin, which is in accordance with the findings of Harris and Chaumba [3]; Pronost, Harris and Pin [6]; and McDonald and Holwell [28]. Fluid-precipitated dolomite, and (pervasive) serpentinization and chloritization, suggest a subsolidus (hydrothermal) fluid alteration of the clinopyroxenite. As revealed by the petrographical and sulphur and Sm–Nd isotope evidence, the sulphur-saturation, as well as the formation of an orthopyroxene-bearing melt, was established before emplacement, and subsequently formed a heterogeneous orthomagmatic

PGE–BMS-assemblage with evidence of crustal contamination in an early stage chamber [3,6,56,76]. This is further supported by the PPGE (Rh, Pt, Pd) concentrations in the BMS and mass-balance calculations, which are in general agreement with those of the Merensky reef and the UG-2, in both of which large proportions of the whole-rock Pd and Rh are hosted by pentlandite, whereas Pt and the IPGE were interpreted to mainly occur in discrete PGM, rather than being hosted by BMS. Pentlandite is a major carrier of Pd in the Platreef clinopyroxenite, which is in agreement with the observations of Junge, Oberthür, Kraemer, Melcher, Piña, Derrey, Manyeruke and Strauss [17] and Klemd, Herderich, Junge, Oberthür, Schouwstra and Roberts [12]. However, the presence of pentlandite with variable PGE concentrations on the thin section scale may be related to variations in the S content at S-saturation, resulting in relatively diluted PGE concentrations during abundant formation of BMS at high sulphur saturation (cf. [38]). Thus, the variations of pentlandite grain PGE concentrations on a thin section scale could be due to the presence of different generations of pentlandite. In addition, as indicated by the sulphide reaction-textures (i.e., their replacement by the late secondary carbonate and serpentine, as well as dissolution along the grain boundaries), mobilization and redistribution may have contributed to the generation of the different pentlandite grain PGE populations.

**Author Contributions:** Conceptualization, R.K.; Data curation, M.K., R.K. and A.B.; Investigation, M.K., A.B.; L.M.; R.K. and M.J.; Methodology, R.K., M.K., A.B., L.M., M.R. and M.J.; Validation, R.K. and A.B.; Writing—original draft, R.K., A.B., and M.K. Writing—review & editing, R.K., A.B., M.K., M.J., L.M. and R.S. All authors have read and agreed to the published version of the manuscript.

**Funding:** This research received no external funding.

**Acknowledgments:** Anglo American Platinum is thanked for access to the drill core and the logistic support. The authors would like to thank Drs. Helene Brätz, Thomas Günther and Sönke Brandt for assisting in LA-ICP-MS analyses and EMPA at the GeoZentrum Nordbayern. We further appreciate the reviews of the two journal referees. Furthermore, the authors acknowledge the facilities, and the scientific and technical assistance of the Australian Microscopy & Microanalysis Research Facility at the Centre for Microscopy, Characterization & Analysis (CMCA), The University of Western Australia, a facility funded by the University, State and Commonwealth Governments. Moreover, special thanks go to Michael Joachimski for providing oxygen- and carbon-isotope analyses.

**Conflicts of Interest:** The authors declare no conflict of interest.

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