

Gasaat Phosphate Project

Major breakthrough as metallurgical tests deliver premium phosphate concentrate

The exceptional results reveal the potential for Gasaat's product to attract a significant price premium

HIGHLIGHTS

- Metallurgical tests show that a simple, single-stage flotation circuit produces commercial high-grade phosphate of up to 34% P_2O_5 with low impurities.
- P_2O_5 concentrate grades are the highest ever reported for the Gasaat Phosphate Project.
- Subject to confirmatory test work, these results open the possibility of PhosCo targeting premium product specifications to attract premium prices with the gap between low grade and high grade prices often being very wide.
- This breakthrough verifies the reduction in process complexity and technical risk announced previously.
- To allow time for some extra test work and to optimise the project economics for these results, the updated scoping study is now expected to be released in October 2026.
- Ore variability testwork to proceed shortly to confirm the feed and concentrate grade limits across KM and other Gasaat deposits.

PhosCo Managing Director, Taz Aldaoud said:

"These results are a major breakthrough for PhosCo because they highlight the potential for our product to attract a substantial price premium. This top-quality product would be highly sought after. This is shown by the fact that phosphate rock with similar characteristics trades at a material premium to regional benchmarks due to its scarcity and cost benefits for transport and downstream processing."

“The ability to produce a desirable phosphate concentrate with a simple flowsheet is a major breakthrough at a time when the world is looking for new suppliers. PhosCo remains firmly on track to become a cost-competitive, globally significant fertiliser producer”.

Gasaat Metallurgical Testwork Results

PhosCo (ASX: PHO) is pleased to announce MgO optimisation metallurgical testwork results for the KM prospect within the Company’s wholly owned Gasaat Phosphate Project in Northern Tunisia. A summary of the flotation test results, carried out at Bureau Veritas Minerals (Adelaide) is listed in Table 1.

Table 1: KM MgO Optimisation Flotation Test Results

Flotation Test	ROM Feed Grade P ₂ O ₅ (%)	Deslimed Feed Grade P ₂ O ₅ (%)	Cleaner Concentrate Grades (%)			Cleaner Recovery P ₂ O ₅ (%)	MER
			P ₂ O ₅	MgO	SiO ₂		
1 kg/t new MgO collector	27.3	27.6	33.0	1.32	3.20	77.8	0.057
1.1 kg/t new MgO collector	27.2	27.6	33.5	1.21	2.52	76.7	0.052
1.3 kg/t new MgO collector	27.2	27.5	34.0	0.96	2.57	73.5	0.044
1 kg/t new MgO collector with re-cleaner	27.2	27.6	33.6	1.18	2.42	74.4	0.052
1 kg/t original MgO collector with re-cleaner	26.4	26.6	31.4	1.35	2.54	73.8	0.059

The results in Table 1 show that the concentrate grades from the flotation tests ranged from 31.4% to 34% P₂O₅ at recoveries between 73.5% to 77.8% P₂O₅. In particular, four concentrate grade results of 33.0%, 33.5%, 33.6% and 34% P₂O₅ are the highest ever achieved for the Gasaat project. The targeted MgO concentrate grade of <1% was attained using a new MgO collector at 1.3 kg/t, which gave a final MgO grade of 0.96%. Each concentrate demonstrated very low SiO₂ contents between 2.42% and 3.2%, whilst the Minor Element Ratios (MER) of between 0.044 and 0.059 ((%Fe₂O₃+% Al₂O₃+% MgO)/(% P₂O₅)) are the lowest ever recorded for Gasaat.

The data in Table 1 demonstrates that the re-cleaning tests using the new and original MgO collectors didn't add any significant upgrade in P₂O₅ or MgO concentrations, compared to the simple rougher-cleaner circuit configuration, so most likely will not be required in the process flowsheet. The data also indicate that in each test there is a slight upgrade in P₂O₅ feed grade of between 0.2 to 0.4%, via desliming the -20 µm fraction, which helps the flotation process upgrade the P₂O₅ to the highest concentrate grades.

Further testing is currently underway to confirm MgO concentrate grades of <1% and P₂O₅ grades >32% can be achieved from lower P₂O₅ feed grades.

Phosphate Market Pricing

The testwork results confirm the potential to produce a premium grade phosphate concentrate of up to 34% P_2O_5 (well above the generally traded grade of 29-32% P_2O_5 used in phosphate fertiliser production) using a simple, proven beneficiation technique.

The phosphate market is very large market with approximately 200MT of rock phosphate currently mined and traded annually, with the export market comprises around 33MT.

Higher grade phosphate is also traded at ~33-34% P_2O_5 often at a significant premium to benchmark prices.

Linear Relationship between MgO Collector Dosage and MgO Concentrate Grade

A linear relationship has been established between the cleaner concentrate MgO grade and the new MgO collector dosage (Figure 1). The straight line equation indicates the 1% MgO concentrate grade target is met when Y (MgO collector dosage) = 1.27 kg/t. The relationship allows lower MgO concentrate grades to be targeted, if required, according to customer final grade specifications and sale price of the concentrate.

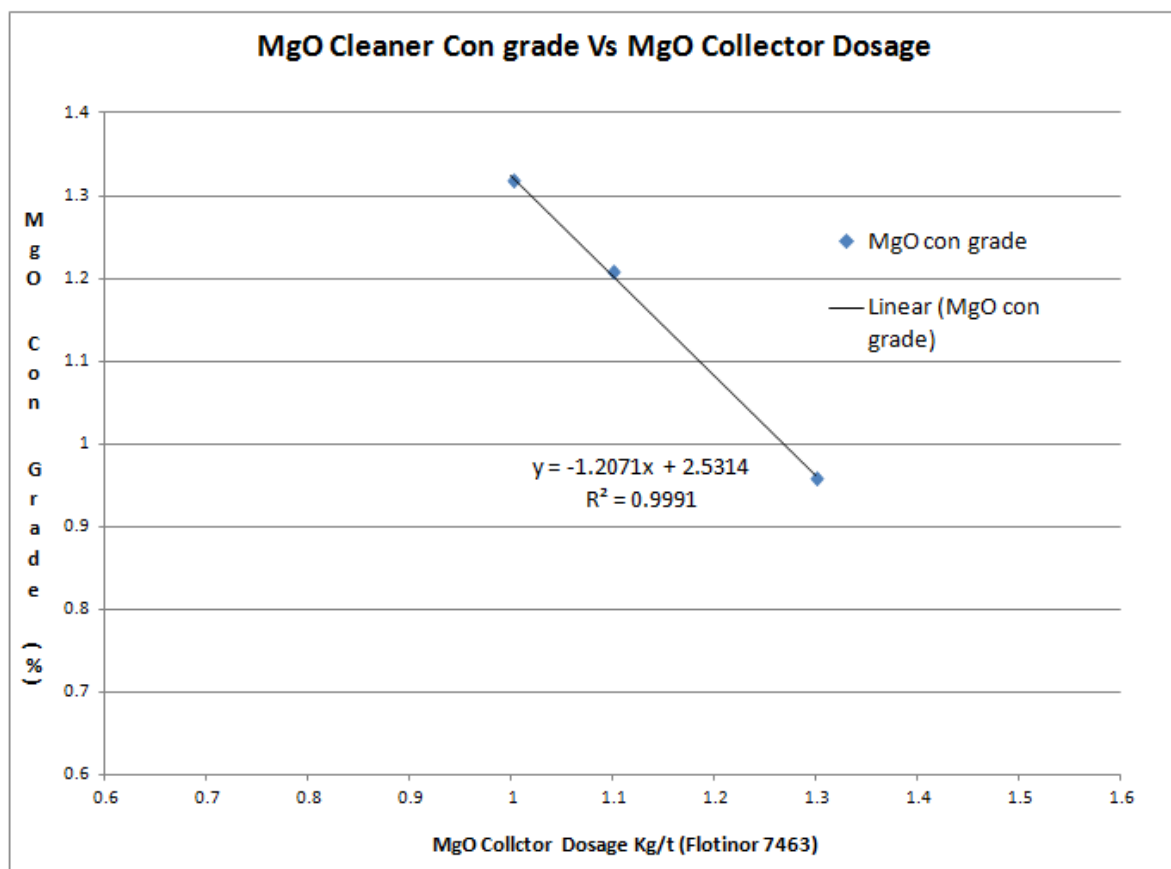


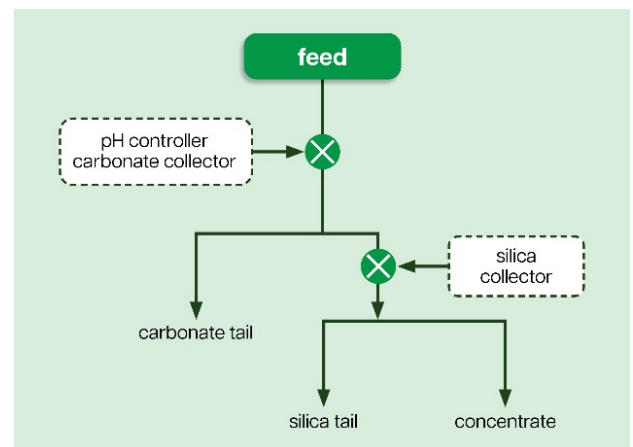
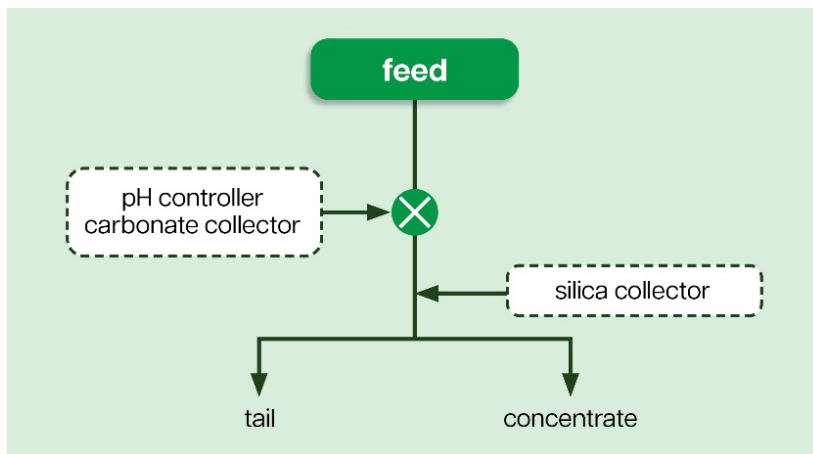
Figure 1. MgO Collector Dosage Versus Cleaner Concentrate MgO Grade.

Verification of Simplified Beneficiation Process

The current MgO optimisation test results, importantly, verify the PhosCo simplified beneficiation process that has been developed during 2026. The results confirm that very high P_2O_5 concentrate grades (33 to 34%), low MgO concentrate grades (<1%) and very low MER (0.044) can be achieved from a simple crush, grind, rougher-cleaner flotation process configuration. It is expected that this simple process will be used going forward to assess ore variability across the KM deposit, along with other Gasaat ores.

As announced previously, the flotation tests carried out on KM phosphate layer B, have established a simplified flowsheet compared to the original Jacobs Engineering Process that was based on a composite of the three phosphate layers of the deposit (A, B and C).

Essentially the new Single Stage flotation process applies the same key processes as Two Stage flotation, just with a materially simplified flowsheet. The proposed PhosCo Single Stage flotation process for Gasaat is a direct reverse flotation method whereby the carbonates and silica are removed simultaneously in a Single Rougher and Cleaner Flotation Stage (see Figures 2a and 2b).



Figures 2a and 2b. Schematic Single Stage Flotation versus Two Stage Flotation

The PhosCo Gasaat simplified beneficiation process flowsheet is shown below in Figure 3 and consists of crushing, attritioning, grinding, single stage rougher and cleaner flotation, dewatering and concentrate drying.

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graph TD
    ROM[ROM Ore] --> Crushing[Primary, Secondary, Tertiary Crushing  
-4mm product]
    Crushing --> Attrition[Attritioning  
60% solids]
    Attrition -- "CUF +20µm" --> RodMill[Rod Mill]
    Attrition -- "COF -20µm" --> Thicker1[Thicker]
    RodMill --> Hydro1[Hydrocyclone]
    Hydro1 -- "CUF +150µm" --> RodMill
    Hydro1 -- "COF P90 150µm" --> Thicker1
    Thicker1 --> BeltFilter1[Belt Filter  
15-20% Moisture]
    BeltFilter1 -- "U/F" --> RotaryDryer[Rotary Dryer  
~5-6% moisture]
    RotaryDryer --> Stockyard[Final Concentrate Stockyard and truck loadout]
    Thicker1 -- "U/F" --> Thicker2[Thicker]
    Thicker2 -- "O/F" --> ProcessWater[Process Water Collection Tank]
    ProcessWater -- "O/F" --> BeltFilter2[Belt Filter  
15-20% Moisture]
    BeltFilter2 -- "U/F" --> ShedDrying[Shed Drying]
    ShedDrying --> DryStackedTail[Dry Stacked Tail]
    DryStackedTail --> Stockyard
    
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Figure 3. Simplified Gasaat Beneficiation Process Flowsheet

This refinement to the previous flowsheet leverages extensive test work on metallurgical and phosphate recovery methodologies at Gasaat has been ongoing for since 2012. Initially by Chaketma Phosphates SA personnel and then in 2016/2017 by enlisting the services of the Jacobs Engineering Group to undertake a feasibility study evaluation to international standards of the process parameters and equipment selection. This cumulated in 2017 with a process design based on pilot metallurgical test work carried out by Jacobs Engineering in Florida on the 28-tonne bulk sample confirming the viability of the reverse flotation flowsheet, including rougher and cleaner carbonate flotation along with rougher and cleaner silica flotation. The 2016/2017 test work used Florida tap water.

An analysis of the previous metallurgical test work is detailed in Section 4 of Company's announcement on 9 December 2022 titled 'Scoping Study Confirms Outstanding Economics for Chaketma.'

Next Steps

- Further MgO optimisation metallurgical testwork.
- Planning of feasibility study testwork, to include: Variability testing of KM and other Gasaat ores, Engineering Design Criteria, Locked Cycle flotation tests.

This announcement is authorised for release to the market by the Board of Directors of PhosCo Ltd.

For further information, please contact:

Taz Aldaoud
Managing Director
T: +61 473 230 558

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COMPETENT PERSON STATEMENT

Statements contained in this report relating to metallurgy results, scientific evaluation, and potential, are based on information compiled and evaluated by Dr Adam Teague, Director of Adam Teague Consulting is a metallurgical engineer with sufficient relevant experience in phosphate mineralisation and metallurgical testwork being reported on, to qualify as a Competent Person as defined in the Australian Code for Reporting of Identified Mineral resources and Ore reserves (JORC Code 2012). Dr Teague consents to the use of this information in this report in the form and context in which it appears.

Previously Reported Results

There is information in this announcement relating to historic data and Exploration Targets, Exploration Results or Mineral Resources which were previously announced on 15 March 2022, 17 November 2022, 9 December 2022, 3 October 2024, 26 November 2024, 13 January 2025, 11 March 2025, 19 March 2025, 28 July 2025, 10 September 2025, 18 November 2025, 18 December 2025, 28 January 2026, 12 February 2026, 26 March 2026, 7 April 2026, 5 May 2026, 26 May 2026, 3 June 2026, 9 June 2026 and 19 August 2026. Other than as disclosed in those announcements, the Company confirms that it is not aware of any new information or data that materially affects the information included in the original market announcements. The information in this announcement relating to the Company's Scoping Study are extracted from the Company's announcement on 9 December 2022 titled 'Scoping Study Confirms Outstanding Economics for Chaketma'. All material assumptions and technical parameters underpinning the Company's Scoping Study results referred to in this announcement continue to apply and have not materially changed. The Company confirms that the form and context in which the Competent Person's findings are presented have not been materially modified from the original market announcement.

Table 1: Detailed Phosphate Recovery and Concentrate Grades for KM MgO Optimisation Flotation Tests

		Recovery		Concentrate Grade					
Sample	Product	P ₂ O ₅ (%)	P ₂ O ₅ (%)	Fe ₂ O ₃ (%)	Al ₂ O ₃ (%)	MgO (%)	CaO (%)	SiO ₂ (%)	MER
Beneficiation Results									
Hole 05 Layer B	Head	100	27.6	0.38	0.60	2.22	46.1	10.8	
1 kg/t New MgO Collector	Cl Tail	77.8	33.0	0.24	0.31	1.32	51.8	3.20	0.057
	Cl Con	2.1	17.5	0.73	1.21	2.77	33.3	31.3	
	Ro Con	20.1	17.6	0.63	1.14	4.02	35.7	24.4	
Hole 05 Layer B	Head	100	27.6	0.38	0.58	2.20	46.1	10.8	
1.1 kg/t New MgO Collector	Cl Tail	76.7	33.5	0.24	0.29	1.21	52.3	2.52	0.052
	Cl Con	3.2	14.7	0.66	0.96	4.16	32.8	30.0	
	Ro Con	20.0	18.0	0.60	1.11	3.86	35.9	23.9	
Hole 05 Layer B	Head	100	27.5	0.39	0.58	2.22	46.1	10.9	
1.3 kg/t New MgO Collector	Cl Tail	73.5	34.0	0.26	0.28	0.96	52.7	2.57	0.044
	Cl Con	5.4	16.4	0.53	0.86	4.17	34.5	27.4	
	Ro Con	21.1	18.5	0.59	1.07	4.03	36.9	21.9	

Hole 05 Layer B	Head	100	27.6	0.39	0.58	2.20	46.2	10.7	
1 kg/t New MgO Collector	Re-Cl Tail	74.4	33.6	0.27	0.29	1.18	52.4	2.42	0.052
Re-Cleaner	Re-Cl Con	0.80	30.0	0.81	1.15	1.70	48.2	6.68	
	Cl Con	15.9	15.9	0.60	0.97	4.15	34.3	27.3	
	Ro Con	18.4	18.4	0.57	1.06	3.80	36.5	23.5	
Hole 05 Layer B	Head	100	26.6	0.37	0.57	2.16	44.7	10.6	
1 kg/t Original MgO Collector	Re-Cl Tail	73.8	31.4	0.24	0.26	1.35	49.6	2.54	0.059
Re-Cleaner	Re-Cl Con	1.20	28.4	0.80	1.11	2.00	46.4	9.64	
	Cl Con	3.90	13.9	0.59	0.95	3.60	30.6	35.1	
	Ro Con	21.1	19.5	0.57	1.13	3.56	37.5	22.0	

JORC Code, 2012 Edition – Table 1

Section 1 Sampling Techniques and Data

(Criteria in this section apply to all succeeding sections.)

Criteria	JORC Code explanation	Commentary
Sampling techniques	■ Nature and quality of sampling (e.g. cut channels, random chips, or specific specialised industry standard measurement tools appropriate to the minerals under investigation, such as downhole gamma sondes, or handheld XRF instruments, etc.). These examples should not be taken as limiting the broad meaning of sampling. ■ Include reference to measures taken to ensure sample representivity and the appropriate calibration of any measurement tools or systems used. ■ Aspects of the determination of mineralisation that are Material to the Public Report. ■ In cases where ‘industry standard’ work has been done, this would be relatively simple (e.g. ‘reverse circulation drilling was used to obtain 1 m samples from which 3 kg was pulverised to produce a 30 g charge for fire assay’). In other cases, more explanation may be required, such as where there is coarse gold that has inherent sampling problems. Unusual commodities or mineralisation types (e.g. submarine nodules) may warrant disclosure of detailed information.	■ Geological information and metallurgical testwork samples referred to in this ASX announcement were derived from diamond drilling programs. Two modes of sampling contribute to the Kodiet Mssafin (KM) Mineral Resource, both completed in the 2025 campaign by Himilco Pty Ltd (a wholly owned subsidiary of PhosCo Ltd): – 13 HQ-diameter (63.5 mm) diamond drill holes (GADD-2025 series, 818.0 m total) – 5 surface channel trenches (GAT-01 to GAT-05). ■ The dataset comprises 483 laboratory major-oxide records covering nine drill holes and all five trenches; an additional 118 portable x-ray fluorescence (pXRF) records cover three drill holes (GADD-2025-28, GADD-2025-29 and GADD-2025-30) where laboratory assays are deferred to the standard core-sampling step. ■ Sample types comprise 319 half-core and 164 full-core records, all collected at nominal 1 m length (mean 0.98 m, range 0.4–1.5 m), to lithological contacts. ■ HQ core was half-cored using a diamond saw; samples were bagged and dispatched to ALS Spain where the entire submitted sample was crushed to –2 mm. ■ Trench sampling used continuous channels (~20 kg/m, average 0.98 m bin length); trenches were cut perpendicular to stratigraphy, corrected for slope and surveyed by differential GPS (DGPS). ■ All material mineralisation is hosted within the Chouabine Formation, which forms a single, laterally continuous four-unit mineralised package: CLP (upper transitional phosphate), CDP (coarse coprolitic phospharudite), CAP (apatitic phospharenite – primary ore) and CMP (lower phosphatic marl). ■ Composite sample material was blended and homogenised and then sub-sampled for various metallurgical testwork.

Criteria	JORC Code explanation	Commentary
Drilling techniques	■ Drill type (e.g. core, reverse circulation, open-hole hammer, rotary air blast, auger, Bangka, sonic, etc.) and details (e.g. core diameter, triple or standard tube, depth of diamond tails, face-sampling bit or other type, whether core is oriented and if so, by what method, etc.).	■ All KM drilling is HQ-diameter diamond core, conducted by Himilco Pty Ltd in the 2025 campaign: – 13 holes totalling 818.0 m, with hole lengths ranging from 31.5–110.0 m (median 60.0 m). ■ The 10 fully processed holes (GADD-2025-03, GADD-2025-05, GADD-2025-06, GADD-2025-07, GADD-2025-08, GADD-2025-11, GADD-2025-12, GADD-2025-13, GADD-2025-14, GADD-2025-15) carry full geological and geotechnical logs. ■ The 3 additional holes (GADD-2025-28, GADD-2025-29, GADD-2025-30) are drilled and pXRF-assayed; geological logging, geotechnical recovery and laboratory assay are deferred to the standard core-sampling step in this campaign workflow. The unsawn core for these three holes was photographically reviewed by the Competent Person and appears competent across all three holes. ■ All holes are essentially vertical: first-station downhole readings span dips of -86.5° to -0.0° across the 10 surveyed holes, with full-hole survey records (7–13 readings per 100 m on the surveyed holes) recording dip range -84.8° to -90.0°. ■ Holes intersect the gently west-dipping mineralised package (10–20° W) at a high angle to bedding (typically 70–80°). ■ Core is not oriented – oriented core is not required for a gently-dipping stratabound target where intercept angle is controlled by bedding geometry.
Drill sample recovery	■ Method of recording and assessing core and chip sample recoveries and results assessed. ■ Measures taken to maximise sample recovery and ensure representative nature of the samples. ■ Whether a relationship exists between sample recovery and grade and whether sample bias may have occurred due to preferential loss/gain of fine/coarse material.	■ Core recovery within the mineralised package (CLP + CDP + CAP + CMP) averages 95.7% by weighted mean across 293.04 m of geotech-logged mineralised intervals. ■ Per-domain recovery is essentially flat across the four Mineral Resource Estimate (MRE) domains: CAP 95.79%, CMP 95.24%, CLP 95.37%, CDP 95.09% – no domain-specific recovery bias has been identified, and no grade-versus-recovery correlation has been identified. ■ CAP, the principal ore domain, accounts for 74% of the mineralised metres logged. ■ Aggregate dataset recovery (mineralised, waste, weathering and unclassified material combined) is 94.7% across 643.90 m.

Criteria	JORC Code explanation	Commentary
Logging	■ Whether core and chip samples have been geologically and geotechnically logged to a level of detail to support appropriate Mineral Resource estimation, mining studies and metallurgical studies. ■ Whether logging is qualitative or quantitative in nature. Core (or costean, channel, etc.) photography. ■ The total length and percentage of the relevant intersections logged.	■ Recovery data are derived from 219 logged intervals across the 10 fully processed drill holes, audited against the SRK interpreted domain model. ■ Six mineralised intervals record sub-90% recovery: two unexplained intervals at GADD-2025-12 in pure CAP rock (a hole-wide mechanical-recovery pattern, with the hole running 91.6% overall); two fault-controlled intervals at GADD-2025-06 (a logged fault 3 m up-hole) and GADD-2025-07 (a complex fault zone west of the main domain); and two marginal intervals at GADD-2025-14 in weak (R1) CMP rock. Net recovery loss across the six intervals is 1.59 m, or 0.5% of mineralised metres logged. ■ No mineralised interval records over 100% recovery. ■ The three additional drill holes (GADD-2025-28, GADD-2025-29 and GADD-2025-30) carry no geotechnical log; recovery will be logged at the standard core-sampling step. The unsawn core was photographically reviewed by the Competent Person and appears competent across all three holes. ■ Recovery is not recorded for the trench samples. ■ Geological logging covers the 10 fully processed drill holes and all 5 trenches: 67 intervals at the major-lithology level across 643.7 m, recording lithology, sub-lithology, grain size where observed, and a free-text description. ■ Estimation domains for the Mineral Resource are derived from the SRK-interpreted lithology dataset – 177 intervals across all 13 drill holes and 5 trenches, refining the field log to a seven-unit stratigraphy (CLN dolomitic nummulitic limestone hanging wall, CLC dolomitic limestone, CLP, CDP, CAP, CMP, CMR mudstone footwall) that supports the four-unit mineralised package definition (CLP + CDP + CAP + CMP). ■ Field-stage pXRF point measurements verify lithological logs; pXRF effectively discriminates the dolomitic limestone caprock, the phosphorite sub-units and the Cretaceous mudstone footwall. ■ Logged lithology is cross-verified against laboratory major-oxide chemistry on receipt of assays, providing geochemical Quality Assurance / Quality Control (QAQC) of the field log and improving

Criteria	JORC Code explanation	Commentary
		the consistency of lithostratigraphic domains in the Mineral Resource. ■ Geotechnical logging covers 219 intervals across the 10 fully processed drill holes (643.90 m), recording recovery, rock quality designation (RQD), rock strength (R1–R3), deformation style and deformation intensity. Strength codes are dominated by R2 (52%) and R3 (36%). ■ Structural logging covers 117 point observations across the same 10 drill holes – fractures (65), veins (33), faults (10) and karst (9). Alpha angle (between structure plane and core axis) is recorded on 87% of records; beta angle is null on all records, as expected for unoriented core. ■ Core boxes are marked with box number, core depths, driller's blocks and sample depths; core has been photographed. ■ The three additional drill holes (GADD-2025-28, GADD-2025-29 and GADD-2025-30) carry no geological, geotechnical or structural log; all three datasets will be logged at the standard core-sampling step.
Sub-sampling techniques and sample preparation	■ If core, whether cut or sawn and whether quarter, half or all core taken. ■ If non-core, whether riffled, tube sampled, rotary split, etc. and whether sampled wet or dry. ■ For all sample types, the nature, quality and appropriateness of the sample preparation technique. ■ Quality control procedures adopted for all sub-sampling stages to maximise representivity of samples. ■ Measures taken to ensure that the sampling is representative of the in situ material collected, including for instance results for field duplicate/second-half sampling. ■ Whether sample sizes are appropriate to the grain size of the material being sampled.	■ HQ core was half-cored using a diamond saw, with sampling guided by pXRF point measurements taken at intervals down each metre. Cutting commences in low-grade overburden several metres above the mineralised package to allow for mining dilution and continues through the phosphate sequence to the underlying mudstone footwall. Samples are taken to the closest lithological boundary, then in increments of approximately 1 m within uniform lithology. ■ Trench channel samples (~20 kg/m) follow the same nominal 1 m sample length and to lithological-boundary scheme. ■ All samples were bagged and dispatched to ALS Spain for preparation and analysis. The entire submitted sample was crushed to -2 mm using a crusher/rotary splitter combination (70% passing 2 mm); a 250 g split is then pulverised to better than 85% passing 75 µm; pulps are sealed in double heat-sealed, air-evacuated plastic bags. ■ Sample types comprise 319 half-core records and 164 full-core records (predominantly trench channels). Sample interval averages 0.98 m (median 1.00 m, range 0.4–1.5 m).

Criteria	JORC Code explanation	Commentary
		■ Sample grain size at KM ranges from coarse coprolitic phospharudite (CDP) through medium- to fine-grained phospharenite (CAP) to micritic marl (CMP); the 1 m nominal sample length is representative of this style of stratabound sedimentary phosphorite. ■ QAQC samples were inserted into the laboratory analytical run: 24 certified reference materials (CRMs – GPO13, GPO14 and GPO17 in approximately equal proportions; 4.97% insertion rate) and 18 coarse blanks (3.73% insertion rate); both within typical 3–5% targets. Three field duplicate pairs (0.62% insertion rate) are recorded, materially below the typical 5–10% target. Composited sample material was blended and homogenised and then subsampled for beneficiation testwork. Beneficiation products were sub-sampled by Bureau Veritas Minerals and submitted for whole ore flotation testwork. Flotation feed, concentrate and tailing samples were submitted and analysed by Bureau Veritas minerals for P₂O₅, Fe₂O₃, Al₂O₃, CaO, MgO and SiO₂ analysis by XRF.
Quality of assay data and laboratory tests	■ The nature, quality and appropriateness of the assaying and laboratory procedures used and whether the technique is considered partial or total. ■ For geophysical tools, spectrometers, handheld XRF instruments, etc., the parameters used in determining the analysis including instrument make and model, reading times, calibrations factors applied and their derivation, etc. ■ Nature of quality control procedures adopted (e.g. standards, blanks, duplicates, external laboratory checks) and whether acceptable levels of accuracy (i.e. lack of bias) and precision have been established.	■ All results were derived from samples submitted to Bureau Veritas Minerals (Adelaide) for geochemical analysis by fused-bead XRF. ■ Major oxides (Al₂O₃, CaO, Fe₂O₃, K₂O, MgO, P₂O₅, SiO₂, LOI) were analysed by ME-XRFO6m – lithium-metaborate fusion followed by XRF – a total-extraction method appropriate for phosphorite chemistry. ■ Standard laboratory QAQC was undertaken and monitored by the laboratory and mass balances for each test reported by Bureau Veritas were reconciled against the feed grade. This is subsequently reviewed by PhosCo on receipt of results.
Verification of sampling and assaying	■ The verification of significant intersections by either independent or alternative company personnel. ■ The use of twinned holes. ■ Documentation of primary data, data entry procedures, data verification, data storage (physical and electronic) protocols. ■ Discuss any adjustment to assay data.	■ Verification of the KM exploration database is provided through an SRK data audit covering collar, survey, geology, geotechnical, structural, laboratory assay, pXRF, density and QAQC tables. ■ Not applicable as it is not relevant to the metallurgical test work referred to in this announcement. ■ Test work data is provided by Bureau Veritas in excel spreadsheets and this is loaded into PhosCo's internal metallurgy database.

Criteria	JORC Code explanation	Commentary
Location of data points	■ Accuracy and quality of surveys used to locate drill holes (collar and down-hole surveys), trenches, mine workings and other locations used in Mineral Resource estimation. ■ Specification of the grid system used. ■ Quality and adequacy of topographic control.	■ No adjustment to assay data was carried out. ■ All 10 fully processed drill hole collars are surveyed by DGPS. The 3 additional drill holes (GADD-2025-28, GADD-2025-29 and GADD-2025-30) carry handheld-GPS coordinates pending DGPS upgrade. ■ All 5 trench collars are DGPS-surveyed. ■ All coordinates are in Universal Transverse Mercator (UTM) Zone 32N, World Geodetic System 1984 (WGS84) datum (European Petroleum Survey Group code EPSG:32632); elevations are in metres above mean sea level. ■ Topographic control derives from an airborne LiDAR survey accurate to approximately ± 0.3 m, supplemented by DGPS outcrop-contact traverses for surface geological mapping. The native LiDAR point cloud was resampled to a 10 m grid digital elevation model (DEM) for the Mineral Resource model. ■ Downhole deviation surveys are available for all 10 fully processed drill holes (76 station records taken at nominal 10 m intervals – 7 to 13 records per 100 m of hole). The 3 additional drill holes carry a single placeholder survey record at depth 0 m and will be downhole-surveyed at the standard core-sampling step.
Data spacing and distribution	■ Data spacing for reporting of Exploration Results. ■ Whether the data spacing and distribution is sufficient to establish the degree of geological and grade continuity appropriate for the Mineral Resource and Ore Reserve estimation procedure(s) and classifications applied. ■ Whether sample compositing has been applied.	■ Drill hole spacing within the KM deposit ranges from approximately 40 m to 50 m in the more densely drilled portions of the deposit to +100 m at the deposit margins; the dataset comprises 13 drill holes and 5 trenches across a surface extent of approximately 579 m east–west by 441 m north–south. ■ Not applicable as this announcement relates to metallurgical testwork. ■ Composites for metallurgical testwork have been created, representing various mineralised domains within the KM deposit.
Orientation of data in relation to geological structure	■ Whether the orientation of sampling achieves unbiased sampling of possible structures and the extent to which this is known, considering the deposit type. ■ If the relationship between the drilling orientation and the orientation of key mineralised structures is considered to have introduced a sampling bias, this should be assessed and reported if material.	■ The KM mineralised horizon is a tabular, gently west-dipping (10–20° W) stratabound phosphorite hosted in a single fault-bounded domain. ■ Drilling comprises 13 essentially vertical diamond drill holes (full-hole dip range -84.8° to -90.0°). Channel trenches were cut perpendicular to stratigraphy and corrected for slope during sample collection.

Criteria	JORC Code explanation	Commentary
Sample security	■ The measures taken to ensure sample security.	■ Vertical drill holes intersect the gently west-dipping bedding at a high angle (approximately 70–80° to the bedding plane), giving apparent downhole intercept lengths within ~6% of true thickness. ■ Channel-trench thicknesses represent true thickness. ■ Faults at KM include a north–south striking, east-dipping (~30°) western bounding fault, an east–west trending southern bounding fault, and two east–west internal faults (Fault_3 and Fault_6) which segment the deposit into the Northern, Central and Southern sub-blocks. Internal faults exhibit predominantly strike-slip displacement with a subordinate normal component; closely spaced drilling has confirmed that mineralisation is sharply sheared and cleanly offset across these structures, with no along-fault thickening. ■ No drilling-orientation bias to sampling has been identified. ■ Core is not oriented; alpha angle (between structure plane and core axis) is the primary orientation measurement on structural records. ■ Drill core boxes are transported on the same day to PhosCo’s core processing facility at Rohia under controlled conditions using a 4x4 field transport system under security supervision. ■ Core is in the custody of the drilling contractor until transported to the Rohia facility, at which point control transfers to PhosCo. Core is logged, sampled and photographed at Rohia. ■ Following mechanical sample preparation, samples are split into 800 g to 1 kg sizes and placed in paper bags, that are placed into 50 L plastic barrels for secure transport. ■ Prepared samples are transported under a documented chain of custody to the airport and dispatched to the ALS laboratory in Spain. ■ pXRF field analysis is conducted on a Hitachi X-MET8000 Expert Geo unit at Rohia; data are downloaded daily by a single qualified technician. ■ Database integrity and the single-operator/single-laboratory dataset structure support chain-of-custody confidence.
Audits or reviews	■ The results of any audits or reviews of sampling techniques and data.	■ Audits and reviews supporting the KM Mineral Resource dataset:

Criteria	JORC Code explanation	Commentary
		– SRK 2026 current-program audits – database audit; core recovery audit; specific gravity (SG) (bulk density) audit; geological model workflow; grade validation summary. – SRK 2026 Gassat Geochemical Review – provides the lithogeochemical framework, cross-prospect geochemical profiling and the basis for the SRK-interpreted domain codes used at KM. ■ No external third party audit of the current KM Mineral Resource has been undertaken as at the date of this report.

Section 2 Reporting of Exploration Results

(Criteria listed in section 1 also apply to this section.)

Criteria	JORC Code explanation	Commentary
Mineral tenement and land tenure status	■ Type, reference name/number, location and ownership including agreements or material issues with third parties such as joint ventures, partnerships, overriding royalties, native title interests, historical sites, wilderness or national park and environmental settings. ■ The security of the tenure held at the time of reporting along with any known impediments to obtaining a licence to operate in the area.	■ The Gassat Project (formerly Chaketma) is held 100% by Himilco Pty Ltd, a wholly owned subsidiary of PhosCo Ltd (ASX:PHO). ■ The current exploration permit was granted on 6 March 2025 for a 3-year term, covering approximately 56 km² across eight phosphate prospects: KEL, GSK, SAB, KEA, DOH, GEZ, KH and KM. KM is the northernmost prospect on the permit. ■ The Chaketma exploration licence was first granted in 2010; held by Chaketma Phosphates SA (CPSA – a joint venture of Celamin Ltd and Tunisian Mining Services) from 2011 to 2015; and transferred to Himilco in 2025. ■ The Company and licence are in good standing with the Tunisian Ministry of Industry, Mines and Energy. ■ At this stage of the exploration licence no joint venture, partnership or off-take agreements are in place, and no private net smelter return royalties have been identified. The project is held under an exclusive exploration licence; no mining (exploitation) rights have been granted. ■ Land use across the KM area is predominantly agricultural and pastoral. No known environmental, heritage or land-tenure issues are currently considered blocking factors, subject to confirmation through the required regulatory studies (environmental impact assessment or environmental and social impact assessment (EIA/ESIA), land and heritage assessments). ■ Any future transition to mining will require relevant mining exploitation permits and regulatory approvals under the Tunisian Mining Code.
Exploration done by other parties	■ Acknowledgment and appraisal of exploration by other parties.	■ Modern exploration at KM has been conducted entirely by Himilco/PhosCo in the 2025 campaign; no CPSA-era drilling, trenching or sampling was undertaken at the KM prospect. ■ Earlier regional studies of the Tunisian phosphates by the Centre d'Études et de Recherches des Phosphates Minéraux (CERPHOS) and the Compagnie des Phosphates de Gafsa (CPG) covered the

Criteria	JORC Code explanation	Commentary
		broader Gasaat licence area as part of regional reconnaissance programs; copies of these studies have not been obtained by PhosCo and they do not contribute data to the current Mineral Resource Estimate. ■ No historical workings (open pits, galleries or test trenches) are recorded at KM. ■ No third party exploration datasets, technical reports or standalone studies specific to the KM prospect have been identified or obtained.
Geology	■ Deposit type, geological setting and style of mineralisation.	■ KM is a stratiform Eocene marine phosphorite deposit of the North African phosphogenic province: ■ Mineralisation is hosted in the Chouabine Formation (Late Palaeocene to Early Ypresian) of the Metlaoui Group, deposited in a shallow restricted marine basin around the Kasserine Island palaeohigh under upwelling-driven nutrient supply. ■ Deposit type: marine sedimentary phosphorite of the Gafsa-Metlaoui style. ■ Structural setting: KM sits on the western edge of the Rouhia Basin – a northwest–southeast collapse graben disrupting the northeast–southwest Hercynian trend of the Tunisian Atlas – and forms a single, northeast-trending, fault-bounded graben. ■ The Chouabine Formation at KM is a seven-unit lithostratigraphic package (base to top): CMR mudstone footwall; CMP marly phospharenite; CAP apatitic phospharenite (primary ore); CDP coprolitic phospharudite; CLP phosphatic limestone (transitional upper); CLC dolomitic limestone; CLN dolomitic nummulitic limestone hanging wall. ■ The mineralised package (CLP + CDP + CAP + CMP) has a median preserved thickness of approximately 30 m, reaching a maximum of 56.4 m in GADD-2025-03 – substantially thicker than the other Chaketma prospects. ■ The Mineral Resource is defined within a single coherent fault-bounded domain. The deposit is bounded to the west by an east-dipping (~30°) north–south striking listric normal fault, to the south by an east–west trending fault, and is internally segmented by two east–west faults (Fault_3 and Fault_6) into the Northern, Central

Criteria	JORC Code explanation	Commentary
		and Southern sub-blocks; stratigraphic continuity across the sub-blocks supports modelling and estimation as a single, geometrically uncut volume. ■ Francolite (carbonate-fluorapatite) is the dominant phosphate mineral.
Drill hole Information	■ A summary of all information material to the understanding of the exploration results including a tabulation of the following information for all Material drillholes: – easting and northing of the drillhole collar – elevation or RL (Reduced Level – elevation above sea level in metres) of the drillhole collar – dip and azimuth of the hole – downhole length and interception depth – hole length. ■ If the exclusion of this information is justified on the basis that the information is not Material and this exclusion does not detract from the understanding of the report, the Competent Person should clearly explain why this is the case.	■ The KM dataset comprises 13 diamond drill holes (GADD-2025 series, 818.0 m total) and 5 channel trenches (GAT-01 to GAT-05). ■ All coordinates are in UTM Zone 32N (WGS84; EPSG:32632); elevations are in metres above mean sea level. ■ All 13 drill holes are essentially vertical: full-hole downhole-survey records on the 10 fully processed holes show a dip range of -84.8° to -90.0°. ■ Ten fully processed drill hole collars are surveyed by DGPS; the three additional drill holes (GADD-2025-28, GADD-2025-29 and GADD-2025-30) carry handheld-GPS coordinates pending DGPS upgrade. ■ All five trench collars are DGPS-surveyed. ■ The three additional drill holes carry a placeholder downhole-survey record at depth 0 m only; full downhole survey is scheduled for the standard core-sampling step. ■ A complete drill hole and trench schedule (collar XYZ, total depth, dip, azimuth and analytical-data coverage) is presented in the companion public report.
Data aggregation methods	■ In reporting Exploration Results, weighting averaging techniques, maximum and/or minimum grade truncations (e.g. cutting of high grades) and cut-off grades are usually Material and should be stated. ■ Where aggregate intercepts incorporate short lengths of high grade results and longer lengths of low grade results, the procedure used for such aggregation should be stated and some typical examples of such aggregations should be shown in detail. ■ The assumptions used for any reporting of metal equivalent values should be clearly stated.	■ <i>Not applicable as this announcement is not reporting exploration results.</i> ■ <i>Not applicable as this announcement is not reporting exploration results.</i> ■ <i>Not applicable as this announcement is not reporting exploration results.</i>

Criteria	JORC Code explanation	Commentary
Relationship between mineralisation widths and intercept lengths	■ These relationships are particularly important in the reporting of Exploration Results. ■ If the geometry of the mineralisation with respect to the drillhole angle is known, its nature should be reported. ■ If it is not known and only the downhole lengths are reported, there should be a clear statement to this effect (e.g. 'down hole length, true width not known').	■ <i>Not applicable as this announcement relates to metallurgical testwork.</i> ■ <i>Not applicable as this announcement relates to metallurgical testwork.</i> ■ <i>Not applicable as this announcement relates to metallurgical testwork.</i>
Diagrams	■ Appropriate maps and sections (with scales) and tabulations of intercepts should be included for any significant discovery being reported These should include, but not be limited to a plan view of drillhole collar locations and appropriate sectional views.	■ <i>Refer to the body of this announcement.</i>
Balanced reporting	■ Where comprehensive reporting of all Exploration Results is not practicable, representative reporting of both low and high grades and/or widths should be practiced to avoid misleading reporting of Exploration Results.	■ <i>All results have been balanced and transparently reported.</i>