Supporting information

Access to long-lived room temperature phosphorescence through auration of 2,1,3-benzothiadiazole

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I. General information

All reagents were purchased from Sigma-Aldrich (Merck) or Fluorochem and used without further purification. Solvents used for synthesis were obtained from VWR Chemicals and used without further purification unless otherwise specified. 1,4-Dioxane was dried over activated 4Å molecular sieves. Solvents for photophysical measurements were used without further purification: dichloromethane (Uvasol, Merck/Supelco); anhydrous 2-methyltetrahydrofuran (2-Me-THF) of ≥99% purity (Sigma-Aldrich). These solvents were degassed in a Schlenk flask sealed with greased joints by 4 cycles of freeze-pump-thaw before the photophysical measurements. Thin-layer chromatography (TLC) was carried out using aluminum plates coated with silica gel (60F₂₅₄, Merck) and visualized through exposure to UV light (λ = 254 nm and/or 365 nm). NMR spectra were recorded at 298 K on an Agilent MR400-DD2 (400 MHz) spectrometer with a OneNMR probe at 400 MHz (¹H), 101 MHz (¹³C), 162 MHz (³¹P), or 128.3 MHz (¹¹B). Chemical shifts (δ) are reported in ppm using the residual solvent peak in CDCl₃ ($\delta_{\rm H}$ = 7.26 and $\delta_{\rm C}$ = 77.2 ppm) or C₆D₆ ($\delta_{\rm H}$ = 7.16 and $\delta_{\rm C}$ = 128.1 ppm) as internal reference. ³¹P chemical shifts were referenced via the lock signal to H₃PO₄. ¹¹B chemical shifts were referenced via the lock signal to BF₃•Et₂O. Coupling constants (J) are given in Hz and the apparent resonance multiplicity is reported as s (singlet), d (doublet), t (triplet), q (quartet), or m (multiplet). Assignment of the ¹H and ¹³C spectra were done with the help of ¹H-¹³C HMBC and ¹H-¹³C HSQC experiments. Highresolution mass spectrometry (HRMS) data, (ES-ToF), was determined at the Division of Mass Spectrometry, Department of Chemistry, Imperial College London, UK.

X-ray crystallography

Single crystals of complexes **3a-e** were obtained by slow diffusion of pentane into concentrated CH₂Cl₂ solutions. Measurements were performed using graphite-monochromatized Mo K α radiation at 170 K using a Bruker D8 APEX-II equipped with a CCD camera. The structure was solved by direct methods (SHELXS-2014) and refined by full-matrix least-squares techniques against F2 (SHELXL-2018). The non-hydrogen atoms were refined with anisotropic displacement parameters. The H atoms of the CH₂/CH groups were refined with common isotropic displacement parameters for the H atoms of the same group and idealized geometry. The H atoms of the same group and idealized geometry and the H atoms of the same group and idealized staggered geometry. CCDC 2026788 and 2125287-2125290 contain the supplementary crystallographic data for this paper. The data can be obtained free of charge from The Cambridge Crystallographic Data Centre via www.ccdc.cam.ac.uk/structures.

Steady-state absorption and emission measurements

All the complexes were recrystallized twice before any measurement. This was done by slow diffusion of pentane into concentrated CH₂Cl₂ solutions.

UV/Vis absorption spectra were acquired on a UV-1650PC Shimadzu instrument (measurements in CH₂Cl₂) or on a Varian Cary 5000 spectrophotometer (measurements in 2-Me-THF) at room temperature using quartz cuvettes (10 mm path length). Absorption maxima (λ_{max}) are reported in nm and the molar extinction coefficient (ϵ) in M⁻¹ cm⁻¹. For each compound three data points with known different concentrations were acquired and the measured absorbances (\leq 1) were plotted against the concentrations. The molar extinction coefficients were then determined according to the Lambert-Beer law as the slope of the linear fit.

Steady-state emission and excitation spectra at room temperature were recorded on a FS5 Spectrofluorometer (Edinburgh Instruments) equipped with a SC-20 (thermostatic) sample holder, in a standardized right-angle geometry configuration. For the measurements, excitation and emission slit widths corresponding to spectral resolutions of 4 nm and 1 nm were used, and the integration time was 0.5 s. For the excitation spectrum shown for monitoring at 620 nm, however, the excitation width was lowered to 2 nm since a *ca*. 150 μ M stock solution was used; the integration time was 0.2 s. All data were automatically corrected for detector response using the supplied correction file from the manufacturer by the data collection software, Fluoracle, on the attached computer.

Emission data collected at 298K were obtained from measurements using freeze-pump thaw degassed solvents. Samples were prepared inside an Argon-filled glovebox and the measurements were carried out in Hellma sealable cells (article number: 117100F-10-40) that had their threads lined with PTFE tape before being sealed. The initial solutions of **3a-e** were made by preparing a volumetric solution of the pre-weighed compound and then further diluting it by volume so that all measured solutions were of 15 μ M concentration. Phosphorescence quantum yields were determined by carrying out relative actinometry, *i.e.*, by comparison with the integrated emission intensity of a standard phosphor with known quantum yield, measured under identical conditions. The phosphor used was [Ru(bpy)₃]²⁺ with a reported quantum yield of 0.018 in aerated acetonitrile,¹ and a suitable emission in the red, centred at 620 nm, very similar to the gold complexes. The quantum yields of the samples could thus be determined using the following equation:

$$\phi_x = \phi_s \times \frac{I_x}{I_s} \times \frac{A_s}{A_x} \times \frac{\eta_x}{\eta_s}$$

where ϕ_x is the quantum yield of the sample, ϕ_s is the known quantum yield of the standard, I_x and I_s are the integrated emission intensities of the sample and standard, respectively, and η_x and η_s are the respective refractive indices of the sample and standard solutions, often approximated as those of the solvents for dilution solutions (which is the case here). Finally, A_x and A_s are the absorbances of the sample and standard, typically kept as close as possible at the given excitation wavelength (355 nm in this case). For these measurements, extra care was taken and a calibration curve for the standard fluorophore, [Ru(bpy)_3][PF_6]_2, was made (Figure S1) so that the integrated emission intensity at any given absorbance <0.1 could be estimated for the standard and compared with the sample. The results are shown in Table 2 in the paper.



Figure S1. Calibration curve for $[Ru(bpy)_3][PF_6]_2$ for the determination of the phosphorescence quantum yield of the gold complexes.

Steady-state emission spectra at 77K were collected using a Horiba Jobin Yvon Fluorolog-3 by employing a cold-finger setup, with liquid nitrogen as coolant. Excitation and emission slit widths corresponding to spectral resolutions of 2 nm and 1 nm were used, and the integration time was 0.1 s. All data were corrected for detector response and fluctuation in excitation intensity.

Time-resolved spectroscopic measurements

Time-correlated single photon counting (TCSPC). TCSPC measurements were carried out to determine the fluorescence lifetimes of the gold complexes. The decay traces were recorded using an FS5 fluorometer, with an SC-20 sample holder, as described above. The excitation source was changed to a pulsed sub-ns excitation source, EPLED340. The typical pulse-width was 810 ps, with an average power of 1 μ W at 10 MHz. The detector employed was an air-cooled and stabilized HS-PMT-920. The monitoring wavelength was 425 nm, and a bandpass corresponding

to a spectral width of 10 nm was used. The measurement range was 200 ns and 4096 channels were chosen for the multi-channel analyzer on the collection software, Fluoracle. The soobtained decays were fit using the curve fitting tool in MATLAB R2020b and were amenable to a monoexponential fit.

Time-resolved emission measurements (TRE). Nanonsecond time-resolved emission measurements were made to accurately determine the phosphorescence lifetimes of the gold complexes. A Q-switched Nd:YAG laser (Model NT342B, EKSPLA; FWHM = 8 ns), was used as the excitation source, whose fundamental emission at 1064 nm was frequency tripled to produce 355 nm light. The output was then attenuated using a series of neutral density filters and used to excite the sample. The repetition rate was 10 Hz, and the pump energy was maintained at <0.8 mJ/pulse (±10%) to minimize photodegradation; absorption spectra were measured before and after measurements to confirm sample integrity, and typical differences were less than 5%. The setup utilized a right-angle detection geometry. The LP920 detection system (Edinburgh Instruments) equipped with a photomultiplier tube was used to record the kinetic data, which was collected using the L900 software (which controlled the Tektronix digital oscilloscope) on the connected computer. The decay traces were recorded with a consistent resolution of 5 nm, and a 435 nm cut-off filter was used to minimize contamination from scattered laser light. Every measurement was averaged over 8 shots. The so-obtained decays were binned every ten points, and then fit using the curve fitting tool in MATLAB R2020b and were amenable to a monoexponential fit model.

Computational details

Geometry optimizations without symmetry constraints were carried out with the Gaussian 16 software (revision A.03),² using the hybrid density functional PBE0,³ with a polarized continuum model (PCM) for CH₂Cl₂ (dielectric $\varepsilon = 8.94$).⁴ The 6-31G*⁵⁻⁷ basis set was used for all atoms except Au, which was described by the Stuttgart relativistic pseudopotential and accompanying basis set ECP60MWB.^{8, 9} This combination of functional and basis set has been previously used to investigate structurally similar Au complexes.^{10, 11} Frequency calculations were performed at the same level of theory to ensure that the optimized structures were located at a potential energy surface minimum by the absence of imaginary frequencies. Stability calculations were also performed for all optimized structures to ensure that all wavefunctions were stable. Vertical excitation energies to singlet and triplet excited states were calculated using time-dependent density functional theory (TD-DFT) at the same level of theory with a PCM for CH₂Cl₂.

Electrochemical analysis

Cyclic voltammetry (CV) was performed on an Epsilon potentiometer equipped with an Ag wire reference electrode, a platinum wire counter electrode and a glassy carbon working electrode. All CV experiments were performed on THF solutions comprising of 2.0 mM of analyte (**3a-e**) and n Bu₄NPF₆ (0.1 M) as supporting electrolyte. Ferrocene was used as an internal standard.¹²

II. Synthesis



Figure S2. Structure and numbering of 2,1,3-benzothiadiazole (BTD). ¹H NMR (400 MHz, C₆D₆) δ 7.77 – 7.61 (m, 2H, H4/7), 7.00 – 6.82 (m, 2H, H5/6). ¹³C NMR (101 MHz, C₆D₆) δ 155.2 (C8/9), 129.0 (C4/7), 121.7 (C5/6).

Benzo[c][1,2,5]thiadiazol-4-ylboronic acid (2).



A 50 mL oven-dried two-neck round-bottom flask was loaded with a magnetic stirbar, 4-bromo-2,1,3-benzothiadiazole (2.00 g, 9.30 mmol), KOAc (2.74 g, 27.9 mmol) and $B_2(pin)_2$ (2.83 g, 11.16 mmol). The flask was flushed with Ar and then fitted with a reflux condenser attached to a constant Ar flow. Dry 1,4-dioxane (30 mL)

was added and the suspension was degassed by bubbling Ar through the solution for 30 min. PdCl₂(dppf) (0.408 g, 0.56 mmol, 6.0 mol%) was added in a single portion. The reaction mixture was stirred at reflux for 90 min and then to cooled to rt. The crude mixture was filtered through a layer of Celite. The reaction flask was rinsed with Et₂O (120 mL), which was used to further flush the Celite plug. The organic layer was washed with brine (2 × 80 mL) and H₂O (80 mL), dried (MgSO₄), filtered and concentrated under reduced pressure until 30 mL of solvent remained. The solution was heated to reflux and, under strong stirring, diethanolamine (1.17 g, 11.16 mmol) was added. The reaction was stirred for 30 min, at which point a dark brown oil had formed. The solution was transferred to a separatory funnel and the reaction flask was rinsed out with H₂O (60 mL) until no viscous residue remained in the flask. The aqueous phase was transferred to the separatory funnel and the layers were separated. The organic layer extracted further with H₂O (2 × 60 mL). The combined aqueous layers were acidified with HCl_(aq) (4M) and stirred for 5 min. The resulting precipitate was filtered off and thoroughly rinsed with H₂O. The residue was dissolved in MeOH and the solution was dried under reduced pressure to afford compound 2 as a lightpink solid (1.03 g, 62 %). ¹H NMR (400 MHz, CDCl₃) δ 8.23 (dd, J = 6.5, 1.0 Hz, 1H), 8.13 (dd, J = 8.8, 1.0 Hz, 1H), 7.69 (dd, J = 8.8, 6.5 Hz, 1H), 6.42 (s, 2H); ¹³C NMR (101 MHz, CDCl₃) δ 158.5, 154.3, 138.0, 129.7, 124.2 (The C–B signal was not observed due to quadrupolar relaxation); ¹¹B NMR (128 MHz, CDCl₃) δ 28.7. HR-MS: (ESI-TOF) calcd for [C₆H₅BN₂O₂S + H]⁺ = 181.0238; found 181.0237.

BTD–Au(I)–PPh₃ (3a)



Compound **2** (21.6 mg, 0.12 mmol), chloro(triphenylphosphine)gold(I) (56.4 mg, 0.11 mmol) and KOH (14.8 mg, 0.26 mmol) were stirred in EtOH (1 mL) for 24 hours. The solution was centrifuged, the supernatant was removed, and the solid residue washed with EtOH (3×1 mL). Compound **3a** was obtained as a colourless solid (63 mg, 88 %). ¹H NMR (400 MHz,

C₆D₆) δ 8.08 (ddd, J_{H-H} = 6.33, J_{H-P} = 6.27, J_{H-H} = 1.2 Hz, 1H, H5), 7.90 (ddd, J_{H-H} = 8.7, J_{H-H} = 1.2, J_{H-P} = 0.7 Hz, 1H, H7), 7.59-7.48 (m, 6H, H2'), 7.38 (ddd, J_{H-H} = 8.68, J_{H-H} = 6.33, J_{H-P} = 1.1 Hz, 1H, H6), 6.97 (m, 9H, H3' and H4'); ¹³C NMR (101 MHz, C₆D₆) δ 170.1 (d, J_{C-P} = 116.5 Hz, C4), 165.0 (C9), 155.5 (d, J_{C-P} = 4.1 Hz, C8), 139.2 (C5), 134.7 (d, J_{C-P} = 13.8 Hz, C2'), 131.7 (C4'), 131.2 (d, J_{C-P} = 2.4 Hz, C1'), 129.5 (d, J_{C-P} = 6.7 Hz, C6), 129.3 (d, J_{H-P} = 10.8 Hz, C3'), 118.5(C7); ³¹P NMR (162 MHz, C₆D₆) δ 42.8. HR-MS: (ESI-TOF) calcd. for [C₂₄H₁₈N₂PSAu + H]⁺ = 595.0672; found 595.0667.

BTD–Au(I)–P^tBu₃ (3b)



Compound **2** (22 mg, 0.12 mmol), chloro(tri-*tert*-butylphosphine)gold(I) (43 mg, 0.10 mmol) and KOH (14 mg, 0.25 mmol) were stirred in EtOH (1mL) for 24 hours. The solution was centrifuged, the supernatant was carefully removed, and the solid residue washed with EtOH (3×1 mL). Compound **3b**

was obtained as a colourless solid (51 mg, 78 %). ¹H NMR (400 MHz, C₆D₆) δ 8.03 (ddd, $J_{H-H} = 6.3$, $J_{H-P} = 5.4$, $J_{H-H} = 1.2$ Hz, 1H, H5), 7.88 (ddd, $J_{H-H} = 8.7$, $J_{H-H} = 1.2$ Hz, $J_{H-P} = 1.0$ Hz, 1H, H7), 7.37 (ddd, $J_{H-H} = 8.7$, $J_{H-H} = 6.3$, $J_{H-H} = 6.3$, $J_{H-P} = 1.2$ Hz, 1H, H6), 1.26 (d, $J_{H-P} = 12.8$ Hz, 27H, H2'); ¹³C NMR (126 MHz, C₆D₆) δ 173.2 (d, $J_{C-P} = 106.4$ Hz, C4), 165.3 (C9), 155.6 (d, $J_{C-P} = 4.1$ Hz, C8), 138.7 (C5), 129.5 (d, $J_{C-P} = 6.2$ Hz, C6), 118.0 (d, $J_{C-P} = 1.0$, C7), 38.8 (d, $J_{C-P} = 14.2$ Hz, C1'), 32.3 (d, $J_{C-P} = 4.9$ Hz, C2'); ³¹P NMR (162 MHz, C₆D₆) δ 91.5. HR-MS: (ESI-TOF) calcd. for [C₁₈H₃₀N₂PSAu + H]⁺ = 535.1611; found 535.1620.

BTD–Au(I)–PCy₃ (3c)



Compound **2** (22 mg, 0.12 mmol), chloro(tricyclohexylphosphine)gold(I) (51 mg, 0.10 mmol) and KOH (14 mg, 0.25 mmol) were stirred in EtOH (1mL) for 24 hours. The solution was centrifuged, the supernatant was carefully removed, and the solid residue washed with EtOH (3×1 mL). Compound **3c** was obtained as a colourless solid (56 mg, 92 %). ¹H NMR

(400 MHz, C₆D₆) δ 8.09 (ddd, J_{H-H} = 6.3, J_{H-P} = 6.0, J_{H-H} = 1.2 Hz, 1H, H5), 7.87 (ddd, J_{H-H} = 8.7, J_{H-H} = 1.2, J_{H-P} = 1.0 Hz, 1H, H7), 7.37 (ddd, J_{H-H} = 8.7, J_{H-H} = 6.3, J_{H-P} = 1.2 Hz, 1H, H6), 1.98-1.89 (m, 6H), 1.89-1.80 (m, 3H, H1'), 1.69-1.57 (m, 6H), 1.55-1.35 (m, 9H), 1.05 (m, 9H)

¹³C NMR (101 MHz, C₆D₆) δ 174.3 (d, *J*_{*C-P*} = 110.0 Hz, C4), 165.3 (C9), 155.7 (d, *J*_{*C-P*} = 4.2 Hz, C8), 138.7 (C5), 129.6 (C6), 118.1 (d, *J*_{*C-P*} = 1.3 Hz, C7), 33.6 (d, *J*_{*C-P*} = 24.4 Hz, C1'), 30.9 (d, *J*_{*C-P*} = 1.4 Hz), 27.4 (d, *J*_{*C-P*} = 11.4 Hz), 26.3 (d, *J*_{*C-P*} = 1.4 Hz); ³¹P NMR (162 MHz, C₆D₆) δ 56.5 (s). HR-MS: (ESI-TOF) calcd. for [C₂₄H₃₆N₂PSAu + H]⁺ = 613.2081; found 613.2065

BTD-Au(I)-IPr (3d)



Compound **2** (22 mg, 0.120 mmol), chloro[1,3-bis(2,6-diisopropylphenyl)imidazol-2-ylidene]gold(I) (61 mg, 0.10 mmol) and KOH (14 mg, 0.25 mmol) were stirred in EtOH (1 mL) for 24 hours. The solution was centrifuged, the supernatant was carefully removed, and the solid residue washed with EtOH (3 × 1 mL). Compound **3d** was obtained as a colourless solid (64 mg, 88 %). ¹H NMR (400 MHz, C₆D₆) δ 7.61 (dd, *J* = 8.7,

1.2 Hz, 1H, H7), 7.54 (dd, J = 6.3, 1.2 Hz, 1H, H5), 7.21 (dd, J = 8.4, 7.1 Hz, 2H, H4"), 7.09 (d, J = 7.9 Hz, 4H, H3"), 7.08 (dd, J = 8.7, 6.3 Hz, 1H, H6), 6.36 (s, 2H, H2'), 2.74 (dq, J = 6.9 Hz, 6.8 Hz, 4H, H1"'), 1.59 (d, J = 6.8 Hz, 12H, H2"' or H3"'), 1.14 (d, J = 6.9 Hz, 12H, H2"' or H3"'). ¹³C NMR (101 MHz, C₆D₆) δ 196.7(C1'), 166.9 (C4), 165.6 (C9), 155.1 (C8), 146.0 (C2"), 139.2 (C5), 135.0 (C1"), 130.6 (C4"), 129.0 (C6), 124.2 (C3"), 122.7 (C2'), 116.9 (C7), 29.2 (C1"'), 24.9 (C2"' or C3"'), 24.0 (C2"' or C3"'). HR-MS: (ESI-TOF) calcd. for [C₃₃H₃₉N₄SAu + H]⁺ = 721.2639; found 721.2637

BTD-Au(I)-IMes (3e)



Compound **2** (22 mg, 0.12 mmol), chloro[1,3-bis(2,4,6-trimethylphenyl)imidazol-2-ylidene]gold(I) (54 mg, 0.10 mmol) and KOH (14 mg, 0.25 mmol) were stirred in EtOH (1mL) for 24 hours. The solution was centrifuged, the supernatant was carefully removed, and the solid residue washed with EtOH (3 × 1 mL). Compound **3e** was obtained as a colourless solid (54 mg, 84%). ¹H NMR (400 MHz, C₆D₆) δ 7.62 (dd, *J* = 6.4, 1.2 Hz, 1H, H5), 7.58 (dd, *J* = 8.7, 1.2 Hz, 1H, H7), 7.03 (dd, *J* = 8.7, 6.4 Hz, 1H, H6), 6.72 (s, 4H, H3''), 6.09 (s, 2H, H2'), 2.14 (s, 12H, H1'''), 2.03 (s, 6H, H2'''); ¹³C NMR

(101 MHz, C_6D_6) δ 195.0 (C1'), 167.2 (C8), 166.0 (C9), 155.2 (c4), 139.4 (C2''), 139.2 (C5), 135.7 (C1''), 135.0 (C2''), 129.6 (C3''), 129.2 (C6), 121.7 (C3'), 116.8 (C7) , 21.1 (C2'''), 18.1 (C1'''). HR-MS: (ESI-TOF) calcd. for [$C_{27}H_{27}N_4SAu + H$]⁺ = 637.1700; found 637.1689

<u>Additional information</u>: The complexes **3a**-**e** have sparing solubility in acetonitrile, good solubility in benzene, THF and 2-Me-THF and, high solubility in CH₂Cl₂ and CHCl₃. Solutions of **3a**-**e** are stable indefinitely in the dark at low temperature.

III. Photophysical data



Figure S3. Solid state emission of compound 3d (λ_{ex} = 254 nm).

Compound	λ_{abs} (nm)
3a	306, 314, 351
3b	305, 312, 348
Зс	306, 312, 348
3d	306, 313, 351
Зе	306, 313, 353

Table S1. UV-Vis absorption values for **3a-e** in degassed CH₂Cl₂ (freeze-pump thaw cycling).

 Table S2. Absorption and emission data for compound 3d in solvents of different polarity.

Solvent	λ _{Abs} (nm)	Wavenumber (cm ⁻¹)	λ _{Em} (nm)	Wavenumber (cm ⁻¹)
Hexane	352	28409	432	23148
Toluene	354	28249	444	22523
CH_2CI_2	356	28090	469	21322
2-Me-THF	354	28249	461	21692
Acetonitrile	353	28329	476	21008
MeOH	356	28090	477	20964



Figure S4. Spectroscopic properties of **3a-3e** 2-Me-THF (conc. 15 μ M): Absorption spectra at room temperature (under air atmosphere); Excitation spectra at room temperature in degassed solution (monitored at 620 nm); Emission at room temperature in degassed solution recorded at 540-800nm, λ_{ex} = 355 nm; Emission at 77K under air atmosphere, λ_{ex} = 355 nm.



Figure S5. Time-resolved emission data for **3a-e** (conc. 15 μ M). Phosphorescence monitored at 620 nm, λ_{ex} = 355 nm, bandwidth = 5 nm.



Figure S6. Panels a)-e): Time-correlated single photon counting (TCSPC) data for **3a-3e** monitored at 450 nm (fluorescence) recorded at room temperature in degassed 2-Me-THF (conc. 15 μ M), λ_{ex} = 341.3 nm, emission bandwidth = 10 nm. Panel f): Data for **3a** plotted together with the IRF (solid black line) and the fluorescence decay of BTD only (open pink squares). The latter is coincident with the IRF.

IV. Computational results

Method validation by reproducing reported work.¹⁰



Figure S7. Complex investigated for method validation, **BTD–C=C–Au–PMe**₃ (compound **1a** from reference 10). Cyclohexyl moieties were truncated to methyl groups to expedite computational time.

Table S3. Table S5 from Ref. 10. Vertical excitation energies to singlet and triplet excited states and their corresponding compositions at the optimized S_0 geometry **BTD–C=C–Au–PMe₃** by TD-DFT.¹⁰

Transition	E (cm⁻¹)	λ (nm)	f	Major contribution	Minor contribution
T1	14642	683	0	HOMO → LUMO (89%)	HOMO-4 → LUMO (7%)
S1	25237	396	0.2736	HOMO \rightarrow LUMO (99%)	-
T2	25811	387	0	HOMO-3 \rightarrow LUMO (89%)	HOMO \rightarrow LUMO+1 (6%)
Т3	27681	361	0	HOMO-4 \rightarrow LUMO (43%)	HOMO-3 → LUMO (7%)
				HOMO \rightarrow LUMO+1 (28%)	
T4	28417	352	0	HOMO-1 \rightarrow LUMO (90%)	HOMO-8 → LUMO (3%)
					HOMO-1 \rightarrow LUMO+1 (2%)
S2	29464	339	0	HOMO-1 \rightarrow LUMO (97%)	HOMO-1 \rightarrow LUMO+1 (2%)

Employing computational details from the methods section, **BTD–C=C–Au–PMe**₃ was successfully optimized, yielding a structure with a singlet ground state (S₀). This wavefunction is stable and contains no imaginary frequencies (Lowest frequency: 14.722 cm^{-1}).

Table S4. Vertical excitation energies to singlet and triplet excited states and their corresponding compositions at the optimized S₀ geometry of **BTD–C=C–Au–PMe₃** by TD-DFT. (This work).

Transition	E (cm⁻¹)	λ (nm)	f	Major contribution	Minor contribution
T1	14636	683.27	0	HOMO → LUMO (67%)	HOMO-4 → LUMO (19%)
S1	25211	396.66	0.2686	HOMO \rightarrow LUMO (70%)	-
Т2	25822	387.26	0	HOMO-3 \rightarrow LUMO (67%)	HOMO $ ightarrow$ LUMO+1 (17%)
Т3	27695	361.08	0	HOMO-4 \rightarrow LUMO (47%)	HOMO-3 $ ightarrow$ LUMO (18%)
				HOMO \rightarrow LUMO+1 (37%)	
Τ4	28387	352.28	0	HOMO-1 \rightarrow LUMO (67%)	HOMO-1 \rightarrow LUMO+1 (12%)
S2	29434	339.74	0	HOMO-1 \rightarrow LUMO (70%)	-



Figure S8. Select frontier MO surfaces (Isovalue: 0.02) at the optimized S₀ geometries of BTD–C=C–Au–PMe₃.

Conclusion: Although minor discrepancies were observed in the TD-DFT results (i.e., vertical excitation energies, oscillator strength), the method used in these calculations yielded results that match well to those found in reference 10. The frontier orbitals associated with the predicted transitions also match well to those found in Table S9 of reference 10.

Calculation of BTD-Au-PMe₃:



Cyclohexyl moieties were truncated to methyl groups to expedite computational time.¹⁰ Attempts to optimize the full complex containing PCy_3 lead to results with imaginary frequencies (i.e., the computed structure is not at a potential surface minimum).

Employing computational details from the methods section, **BTD–Au–PMe₃** was successfully optimized, yielding a structure with a singlet groundstate (S_0). This wavefunction is stable and contains no imaginary frequencies (Lowest frequency: 16.398 cm⁻¹).

Table S5. Vertical excitation energies to singlet and triplet excited states and their corresponding compositions at the optimized S₀ geometry of **BTD–Au–PMe₃** by TD-DFT.

Transition	E (cm⁻¹)	λ (nm)	f	Major contribution	Minor contribution
T1	16887	592.16	0	HOMO → LUMO (69%)	HOMO-8 \rightarrow LUMO (12%)
Т2	25990	384.76	0	HOMO-2 \rightarrow LUMO (70%)	
Т3	28566	350.07	0	HOMO-1 \rightarrow LUMO (69%)	HOMO-5 \rightarrow LUMO (12%)
S1	29221	342.22	0.1238	HOMO \rightarrow LUMO (70%)	-
S2	30381	329.15	0.0010	HOMO-1 \rightarrow LUMO (70%)	
T4	32142	311.12	0	HOMO \rightarrow LUMO+1 (41%)	
				HOMO-10 \rightarrow LUMO (34%)	
				HOMO \rightarrow LUMO+6 (32%)	



Figure S9. Selecteded frontier MO surfaces (Isovalue: 0.02) at the optimized S₀ geometries of BTD-Au-PMe₃.

Similar to **BTD–C=C–Au–PMe**₃, the HOMO and LUMO of **BTD–Au–PMe**₃ are predominantly localized on the benzothiadiazole ligand with minimal contribution from the phosphine. Both orbitals are comprised of antibonding combinations of an Au(d) orbital with the π -system of BTD.

Conclusion: Removing the acetylide moiety (**BTD–C≡C–Au–PMe**₃ *vs* **BTD–Au–PMe**₃) yields a 4000 cm⁻¹ blue shift in the calculated vertical excitation energies.

Calculation of BTD-Au-IPr (3d):



Employing computational details from the methods section, BTD–Au–IPr (**3d**) was successfully optimized, yielding a structure with a singlet ground state (S_0). This wavefunction is stable and contains no imaginary frequencies (Lowest frequency: 12.0376 cm⁻¹).

Table S6. Vertical excitation energies to singlet and triplet excited states and their corresponding compositions at the optimized S_0 geometry of BTD-Au-IPr (**3d**) by TD-DFT.

Transition	E (cm ⁻¹)	λ (nm)	f	Major contribution	Minor contribution
T1	16730	597.73	0	HOMO \rightarrow LUMO (0.68)	HOMO-12 → LUMO (0.10)
Т2	26040	384.02	0	HOMO-4 \rightarrow LUMO (0.70)	-
Т3	28343	352.82	0	HOMO-1 \rightarrow LUMO (0.65)	HOMO-7 \rightarrow LUMO (0.15)
Τ4	28500	350.88	0	HOMO-2 → LUMO+5	
Т5	28501	350.86	0		
S1	28702	348.41	0.1289	HOMO \rightarrow LUMO (0.70)	



Figure S10. Selected frontier MO surfaces (Isovalue: 0.02) at the optimized S₀ geometries of BTD–Au–IPr (3d).

Calculation of BTD-Au-IMes:



Employing computational details from the general methods section, **BTD–Au–NHCMe**₃ was successfully optimized, yielding a structure with a singlet ground state (S₀). This wavefunction is stable and contains no imaginary frequencies (Lowest frequency: 5.9913 cm⁻¹).

Table S7. Vertical excitation energies to singlet and triplet excited states and their corresponding compositions at the optimized S₀ geometry of BTD–Au–IMes (**3e**) by TD-DFT.

Transition	E (cm ⁻¹)	λ (nm)	f	Major contribution	Minor contribution
T1	16746	597.16	0	HOMO \rightarrow LUMO (0.68)	HOMO-12 → LUMO (0.13)
T2	26028	384.20	0	HOMO-6 \rightarrow LUMO (0.70)	-
Т3	28399	352.13	0	HOMO-2 → LUMO+2 (0.38)	
T4	28416	351.92	0	HOMO-2 → LUMO+3 (0.37)	
T5	28528	350.53	0	HOMO-1→ LUMO (0.62)	HOMO-7 \rightarrow LUMO (0.17)
S1	28807	347.14	0.1160	HOMO \rightarrow LUMO (0.70)	-



Figure S11. Selected frontier MO surfaces (Isovalue: 0.02) at the optimized S₀ geometries of BTD–Au–IMes (3e).

Conclusion: The predicted vertical excitation energies for BTD–Au–IPr (**3d**) and BTD–Au–IMes (**3e**) occurs between the HOMO and LUMO. Their orbitals are similar in composition to those found in **BTD–Au–PMe₃**, comprising of antibonding combinations of a Au(d) orbital with the π -system of BTD.

V. Electrochemical data



Cyclic voltammograms of BTD–Au–PPh₃:

Figure S12. CVs of **BTD–Au–PPh₃** (**3a**) in THF containing ferrocene internal standard (left) and without internal standard (right).

Table S8. Redox potentials of BTD-Au-PPh₃ (3a) in THF versus Fc⁺/Fc.

Redox Values of 3a			
Event	Potential (V)		
E ₁	0.95		
E ₂	1.05		

Cyclic voltammograms of BTD–Au–P^tBu₃ (3b):



Figure S13. CVs of **BTD–Au–PtBu₃** (**3b**) in THF containing ferrocene internal standard (left) and without internal standard (right).

Table S9. Redox potentials of BTD-Au-PtBu₃ (3b) in THF versus Fc⁺/Fc.

Redox Values of 3b		
Event	Potential (V)	
E ₁	-2.34	
E ₂	1.13	

Cyclic voltammograms of BTD–Au–PCy₃ (3c):



Figure S14. CVs of BTD–Au–PCy₃ (**3c**) in THF containing ferrocene internal standard (left) and without internal standard (right).

Table S10. Redox potentials of BTD–Au–PCy₃ (3c) in THF versus Fc⁺/Fc.

Redox Values of 3c		
Event	Potential (V)	
E1	-2.41	
E ₂	1.11	

Cyclic voltammograms of BTD–Au–IPr (3d):



Figure S15. CVs of BTD–Au–IPr (**3d**) in THF containing ferrocene internal standard (left) and without internal standard (right).

Table S11. Redox potentials of BTD–Au–IPr (**3d**) in THF versus Fc^*/Fc .

Redox Values of 3d		
Event	Potential (V)	
E1	-2.46	
E ₂	1.15	

Cyclic voltammograms of BTD–Au–IMes (3e):



Figure S16. CV of BTD–Au–IPr (**3e**) in THF containing ferrocene internal standard (left) and without internal standard (right).

Table S12. Redox poter	ntials of BTD–Au–IPr (3e	e) in THF versus Fc ⁺ /Fc.
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Redox Values of 3e			
Event	Potential (V)		
E ₁	-2.37		
E ₂	0.93		
E ₃	1.09		

VI. Crystallographic details

The complexes all exhibit linear two-coordinate geometry; the phosphine complexes exhibit C1– Au–P1 angles closer to the ideal arrangement than the C1–Au–C7 angles of the *N*-heterocyclic carbene complexes (177.79(6) – 178.92(10)° versus 166.6(4) – 175.01(13)°), due to the larger steric bulk at the metal centre imposed by the NHC ligands. The Au1–P1, Au1–C1, and Au1–C7 distances are comparable to those of related Au(I) complexes. The Au–C1 and Au–C7 bonds in **3d** are slightly elongated when compared to **3e** (Au–C1: 2.086(4) and 2.052(4) Å; Au–C7: 2.3172(8) and 2.2954(10) Å respectively), likely due to the greater steric demand of the IPr ligand compared to the IMes ligand. The Au–C and Au–P bonds decrease in the order PtBu₃>PCy₃>PPh₃ (Au-C: 2.086(4), 2.052(4), and 2.0455(17) Å; Au-P: 2.3172(8), 2.2954(10), and 2.2845(5) Å, respectively), which is consistent with the decreasing σ -donation strength of the phosphine ligand. The Au–C and Au–P bonds are shortest for the PPh₃ ligand, which is the weakest σ -donating ligand, and strongest pi-acceptor.

Specific for **3a** (PPh₃ ligand): No additional information.

Specific for **3b** (P^tBu₃ ligand): The gold centre, the benzothiadiazole ligand, and one of the phosphine tertiary butyl groups are disordered along a crystallographic mirror plane.

Specific for 3c (PCy₃ ligand): There is a positional disorder of the gold centre and the benzothiadiazole ligand.

Specific for **3d** (IPr ligand): There is a positional disorder of one of the NHC isopropyl groups. Specific for **3e** (IMes ligand): There are two molecules of **3e** in the unit cell. One of these molecules exhibits a positional disorder for the benzothiadiazole ligand over two sites.

CCDC 2026788 and 2125287-2125290 contain the supplementary crystallographic data for this paper. The data can be obtained free of charge from The Cambridge Crystallographic Data Centre via www.ccdc.cam.ac.uk/structures.

Compound	3a (PPh₃)	3b (P ^t Bu ₃)	3c (PCy ₃)	3d (IPr)	3e (IMes)
Chemical formula	C ₂₄ H ₁₈ AuN ₂ PS	$C_{18}H_{30}AuN_2PS$	C ₂₄ H ₃₆ AuN ₂ PS	C ₃₃ H ₃₉ AuN ₄ S	C ₂₇ H ₂₇ AuN ₄ S
<i>M</i> _r	590.40	534.43	612.54	720.71	636.55
Crystal system,	Monoclinic	Monoclinic	Triclinic	Monoclinic	Triclinic
space group	P21/c	P2₁/m	P ⁻ 1	P21/c	P ⁻ 1
Temperature (K)	170	170	170	170	170
	8.4960(4)	8.4072(2)	9.4554(4)	11.819(2)	10.3239(4)
a, b, c (Å)	11.3452(6)	13.1964(4)	11.6097(6)	14.047(3)	15.6203(6)
	21.6600(11)	9.2110(2)	11.8845(5)	18.837(3)	16.6809(7)
	-	-	96.835(1)	-	105.253(1)
a, b, g (°)	97.820(1)	93.753(1)	111.00(3)	96.458(4)	100.318(1)
-	-	-	94.095(1)	-	102.032(1)
V (ų)	2068.37	1019.72(5)	1200.09(10)	3107.5(9)	2458.97(17)
Ζ	4	2	2	4	4
Radiation type	Мо Ка	Мо Ка	Мо Ка	Мо Ка	Мо Ка
m (mm⁻¹)	7.30	7.4	6.3	4.83	6.09
Crystal size (mm)	$0.25 \times 0.20 \times 0.10$	$0.30 \times 0.30 \times 0.07$	$0.6 \times 0.4 \times 0.3$	$0.20 \times 0.09 \times 0.09$	0.22 × 0.19 × 0.12
Diffractometer	Bruker D8 APEX-II	Bruker D8 APEX-II	Bruker D8 APEX-II	Bruker D8 APEX-II	Bruker D8 APEX-II
Absorption correction	Multi-scan	Multi-scan	Multi-scan	Multi-scan	Multi-scan
T _{min} , T _{max}	0.471, 0.746	0.436, 0.747	0.676, 0.749	0.474, 0.746	0.610, 0.746
No. of measured.	59053	41113	53061	120163	133098
independent	5612	4223	6700	7800	13393
and observed [I > 2σ(I)] reflections	5203	3291	6212	6957	11662
R _{int}	0.028	0.054	0.042	0.064	0.037
(sin Θ/λ) _{max} (Å ⁻¹)	0.687	0.782	0.694	0.672	0.691
$R[F^2 > 2s(F^2)],$ $wR(F^2), S$	0.014, 0.032, 1.03	0.030, 0.076, 1.03	0.016, 0.036, 1.06	0.018, 0.044, 1.05	0.024, 0.056, 1.06
No. of parameters	262	150	353	406	665
No. of restraints	0	0	176	39	686
H-atom treatment	Constrained	Constrained	Constrained	Constrained and independent	Constrained
Δ> _{max} , Δ> _{min} (e Å ⁻³)	0.77, -0.35	1.03, -1.02	0.45, -0.66	0.68, -0.70	2.16, -1.69
CCDC No.	2125288	2125287	2026788	2125289	2125290

 Table S13. Crystal data and structure refinement for 3a-e.

Pale yellow crystals of **3a** suitable of single crystal X-ray diffraction were grown by slow diffusion of pentane into a concentrated CH_2Cl_2 solution of **3a**. Compound **3a** crystallises in the monoclinic space group $P2_1/c$. The solid-state structure shows a two coordinate Au centre, coordinated to a phosphine and an aryl ligand. The arrangement is near linear with an angle of 177.79°. The Au-P and Au-C are comparable to other gold(I) aryl phosphine complexes.



Figure S17. Solid-state structure of **3a**. Ellipsoids at 50% probability, methyl groups as capped sticks and hydrogen atoms removed for clarity. Selected metrics: Au1–C1: 2.0455(17) Å; Au1–P1: 2.2845(5) Å; C1–Au1–P1: 177.79(6)°.

Pale yellow crystals of **3b** suitable of single crystal X-ray diffraction were grown by slow diffusion of pentane into a concentrated CH_2Cl_2 solution of **3b**. Compound **3b** crystallises in the monoclinic space group $P2_1/m$. The solid-state structure shows a two coordinate Au centre, coordinated to a phosphine and an aryl ligand. The arrangement is near linear with an angle of 178.92°. The Au-P and Au-C are comparable to other gold(I) aryl phosphine complexes.



Figure S18. Solid-state structure of **3b**. Ellipsoids at 50% probability, methyl groups as capped sticks and hydrogen atoms removed for clarity. Selected metrics: Au1–C1: 2.086(4) Å; Au1-P1: 2.3172(8) Å; C1–Au1–P1: 178.92(10)°.

Pale yellow crystals of **3c** suitable of single crystal X-ray diffraction were grown by slow diffusion of pentane into a concentrated CH₂Cl₂ solution of **3c**. Compound **3c** crystallises in the triclinic space group P–1. The solid-state structure shows a two coordinate Au centre, coordinated to a phosphine and an aryl ligand. The arrangement is near linear with an angle of 177.90°. The Au–P and Au–C are comparable to other gold(I) aryl phosphine complexes.



Figure S19. Solid-state structure of **3c**, (TK1901). Ellipsoids at 50% probability, methyl groups as capped sticks and hydrogen atoms removed for clarity. Selected metrics: Au1–C1: 2.052(4) Å; Au1–P1: 2.2954(10) Å; C1–Au1–P1: 177.90(9)°.

Pale yellow crystals of **3d** suitable of single crystal X-ray diffraction were grown by slow diffusion of pentane into a concentrated CH_2Cl_2 solution of **3d**. Compound **3d** crystallises in the monoclinic space group $P2_1/c$. The solid-state structure shows a two coordinate Au centre, coordinated to a NHC and an aryl ligand. The arrangement is near linear with an angle of 175.01°. The Au-C_{NHC} and Au-C_{aryl} bonds are comparable to other NHC gold(I) aryl complexes.



Figure S20. Solid-state structure of **3d**. Ellipsoids at 50% probability, methyl groups as capped sticks and hydrogen atoms removed for clarity. Selected metrics: Au1–C1: 2.035(2) Å; Au1–C7: 2.0260(18) Å; C1–Au1–C7: 175.01(8)°.

Pale yellow crystals of **3e** suitable of single crystal X-ray diffraction were grown by slow diffusion of pentane into a concentrated CH_2Cl_2 solution of **3e**. Compound **3e** crystallises in the triclinic space group P–1. The solid-state structure shows a two coordinate Au centre, coordinated to a NHC and an aryl ligand. The arrangement is near linear with an angle which ranges 166.64-174.09°. The Au-C_{NHC} and Au-C_{aryl} bonds are comparable to other NHC gold(I) aryl complexes.



Figure S21. Solid-state structure of **3e**. Ellipsoids at 50% probability, methyl groups as capped sticks and hydrogen atoms removed for clarity. Selected metrics: Au1–C1: 2.099(3)-2.128(2) Å; Au1–C7: 2.019(3)-2.020(3) Å; C1–Au1–C7: 166.6(4)-174.09(13)°.

Compound	3a	3b	Зс	3d	Зе
L =	PPh ₃	P^tBu_3	PCy ₃	IPr	IMes
Au-C1	2.0455(17)	2.086(4)	2.052(4)	2.035(2)	2.099(3) – 2.128(2)
Au-P1	2.2845(5)	2.3172(8)	2.2954(10)	-	-
Au-C7	-	-	-	2.0260(18)	2.019(3) — 2.020(3)
C1-Au-P1	177.79(6)	178.92(10)	177.90(9)	-	-
C1-Au-C7	-	-	-	175.01(8)	166.6(4) — 174.09(13)

Table S14. Selected bond lengths (Å) and angles (°) for 3a-e.

VII. NMR Spectra

¹H NMR (400 MHz, C_6D_6) for 2,1,3-benzothiadiazole (BTD).



¹H NMR (400 MHz, CDCl₃) for compound **2**.







¹H NMR (400 MHz, C_6D_6) for compound **3a**.



 ^{13}C NMR (101 MHz, $C_6D_6)$ for compound 3a.



 ^{31}P NMR (162 MHz, $C_6D_6)$ for compound 3a.



¹H NMR (400 MHz, C_6D_6) for compound **3b**.



 ^{13}C NMR (101 MHz, CDCl₃) for compound **3b**.







¹H NMR (400 MHz, C_6D_6) for compound **3c**.



 ^{13}C NMR (101 MHz, $C_6D_6)$ for compound 3c.



 ^{31}P NMR (162 MHz, C₆D₆) for compound **3c**.



^1H NMR (400 MHz, C₆D₆) for compound **3d**.





¹H NMR (400 MHz, C₆D₆) for compound **3e**.



 ^{13}C NMR (101 MHz, C₆D₆) for compound **3e**.



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