Comparison of Basicity of the Diimine and Quinoid Group of 1,10-Phenanthroline-5,6-dione Ligated on Pt(II)

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A platinum(II) complex bearing 1,10-phenanthroline-5,6-dione (phenO₂), [PtCl(phenO₂-N,N)(PPh₃)]⁺ ([2]⁺) was newly prepared and characterized. An X-ray crystallographic study of [2](PF₆) revealed that the phenO₂ ligand coordinates to platinum with the N,N'-chelate form. The basicity of the quinoid part of phenO₂ in [2]⁺ and of the diimine one in [Pt(phenO₂-O,O')(PPh₃)₂] with an O,O'-chelate ring (3) was evaluated in the absence and the presence of Li⁺ by means of UV–vis spectroscopy and cyclic voltammetry. The quinoid group in [2]⁺ practically had no ability to bind Li⁺, but induced on anodic potential shift of the phenO₂ localized redox potential by 850 mV due to the strong interaction between the reduced form of [2]⁺ and Li⁺. Undeniably, a weak interaction of Li⁺ with the diimine moiety of free phenO₂ (K = 3.6 M⁻¹) is greatly enhanced in the interaction between Li⁺ and the diimine site in 3 because of the electron donation from the catecholato group of the complex.

Metal complexes bearing electroactive ligands with two or more accessible oxidation states exhibit unique electronic structures resulting from the combination of the oxidation states of the metal and ligands. Dioxolene principally has an ability to coordinate on metals with three different oxidation states; quinone, semiquinonato, and catecholato forms.¹ Metal-dioxolene complexes often feature delocalization of π electrons over the metal and ligands (charge distribution) due to close energy levels between the d orbital of the central metal and π^* orbital of the ligands.^{2,3} Polypyridyl–metal complexes also show ligand-localized and metal-centered redox reactions.⁴ Intramolecular electron transfer between the two redox sites of metals and ligands often generates characteristic CT bands that are regulated by external stimuli such as photo-irradiation, heat reactions, and redox reactions. Therefore, metal complexes having redox active ligands are feasible candidates for molecular switchings.

The redox behavior of metal–phenO₂ complexes (phenO₂ = 1,10-phenanthroline-5,6-dione) is of interest because phenO₂ has the combined features of polypyridyl and dioxolene ligands. Furthermore, phenO₂ has an ability to coordinate to metals with O,O'-chelate (o-quinoid form) or N,N'-one (diimine form), or both of them as a bridging ligand connecting two metals.^{5–12} Recently, we have reported the redox behavior of Ru(II) complexes having phenO₂.¹³ We herein report the redox behavior of Pt^{II}(phenO₂-N,N') having an N,N'-chelate ring and of Pt^{II}(phenO₂-O,O') with an O,O'-chelate ring to evaluate the basicity of quinoid and diimine groups (Fig. 1).

Results and Discussion

Synthesis and Structure of Platinum(II) Complexes. Treatment of $[PtCl_2(PhCN)_2]$ with $phenO_2$ in dimethyl sulfoxide afforded $[PtCl_2(phenO_2-N,N')] \cdot DMSO$ ([1] $\cdot DMSO$) as yellow needles (Eq. 1).

1,10-phenanthroline-5,6-dione (phenO₂)

$$\begin{bmatrix}
O & N & Pt & Cl \\
O & N & Pt & PPh_3
\end{bmatrix}^+$$

$$[PtCl(phenO2-N,N')(PPh3)]^+$$

$$[2]^+$$

$$Ph_3P & O & N \\
Ph_3P & O &$$

Fig. 1. 1,10-Phenanthroline-5,6-dione (phen O_2) and Pt^{II} –phen O_2 complexes discussed in this study.

$$PtCl_{2}(PhCN)_{2} \xrightarrow{DMSO} O \xrightarrow{N} Pt \xrightarrow{Cl} DMSO$$
 (1)

X-ray crystallography of [1]•DMSO revealed a squareplanar structure of Pt coordinated by two nitrogen atoms of phenO₂ and two chlorine atoms. ¹⁴ The DMSO molecule of [1]•DMSO resides as a solvated molecule in a space in the crystal structure. In fact, the ${}^{1}H$ NMR spectrum of [1]•DMSO in DMSO- d_{6} displayed only three signals for phenO₂ protons, indicating a symmetrical structure of the complex. Thus, the molecular structure of [1]•DMSO is essentially the same as that of [PtCl₂(phenO₂-N,N')] (1). 15,16 The solubility of [1]•DMSO and 1 in organic solvents was too low for measurements of redox behavior of the complexes. Therefore, we introduced PPh₃ into the Pt(phenO₂-N,N') framework to improve the solubility of the complex in ordinary solvents. The reaction of 1 with 1 equiv of PPh₃ in CH₃OH at 50 °C for 1.5 h and the subsequent anion exchange gave [PtCl(phenO₂-N,N')(PPh₃)](SbF₆) ([2](SbF₆)) in moderate yield (Eq. 2).

The ESI-MS spectrum of [2]+ in CH₃CN exhibited the parent peak at m/z 703. The IR spectrum of $[2]^+$ displayed a strong band at 1709 cm⁻¹ assigned to the ν (CO) band. The value is very close to those of 1 (1703 cm⁻¹)¹⁵ and [PtCl₂(Me₂ $phenO_2-N,N')$] (Me₂phenO₂ = 2,9-dimethyl-1,10-phenanthroline-5,6-dione) (1699 cm⁻¹), ¹⁶ and a slightly higher shifted wavenumber compared with free phenO₂ (1675 cm⁻¹).^{5a} The UV-vis spectrum of [2](SbF₆) displayed a band at 310 nm, which can be attributed to the intraligand π - π * transition by analogy with the spectra of $[PtCl_2(bpy)]$ (bpy = 2,2'-bipyridine) (313 and 325 nm in DMF) and $[Pt(bpy)(PMe_3)_2]^{2+}$ (314 and 324 nm in DMF).¹⁷ On the other hand, [Pt(catecholato)-(PPh₃)₂]¹⁸ exhibits a characteristic LMCT band in the region of 400 to 500 nm. The appearance of the $\nu(CO)$ band at 1709 cm⁻¹ in the IR spectrum and the absence of absorption bands in the region of 400 to 500 nm in the electronic absorption spectrum of [2]⁺ indicate that phenO₂ is bound to Pt with the N,N'-chelate.

Figure 2 shows the ORTEP view of [2]⁺ determined by X-ray crystallography. The details of data collection, refinement for the complex, and selected bond lengths and angles are listed in Table 1 and 2, respectively. The Pt atom is coordinated in square-planar form with two nitrogen, one chlorine, and one phosphorus atom. Molecular structures of Pt(II) complexes are subject to the steric hindrance of ligands. For example, two

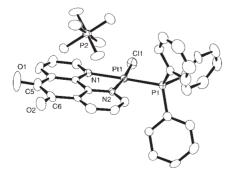


Fig. 2. ORTEP drawing of [2](PF₆) with 50% thermal ellipsoids. Hydrogen atoms are omitted for clarity.

Table 1. Crystallographic Data for Complex [2](PF₆)

	[2](PF ₆)	
Formula	C ₃₀ H ₂₁ ClF ₆ N ₂ O ₂ P ₂ Pt	
Formula weight	847.99	
Color	Pale_Yellow	
Crystal size/mm ³	$0.40 \times 0.10 \times 0.05$	
Crystal system	orthorhombic	
Space group	Pbca	
a/Å	14.391(5)	
$b/\mathrm{\AA}$	14.998(6)	
$c/ ext{Å}$	27.80(1)	
$\alpha/{ m deg}$	90	
$eta/{ m deg}$	90	
γ/deg	90	
$V/{ m \AA}^3$	6000(1)	
Z	8	
T/K	-173	
$D_{ m calcd}/{ m gcm^{-3}}$	1.877	
Radiation	$Mo K\alpha (\lambda = 0.71070 \text{ Å})$	
μ/cm^{-1}	49.24	
F(000)	3280.00	
$2\theta_{ m max}/{ m deg}$	55.0	
No. of reflections	6787	
(all, $2\theta < 54.97^{\circ}$)	0787	
No. of variables	397	
Reflection/Parameter	17.10	
Ratio	17.10	
GOF	1.27	
$R1^{\mathrm{a}}$ $[I > 2\sigma(I)]$	0.056	
R ^{b)} (all data)	0.082	
$R_{\rm w}^{\rm c)}$ (all data)	0.139	

a) $R1 = \Sigma ||F_o| - |F_c||/\Sigma |F_o|$. b) $R = \Sigma (F_o^2 - F_c^2)/\Sigma F_o^2$. c) $R_w = [(\Sigma \omega (F_o^2 - F_c^2)^2/\Sigma \omega (F_o^2)^2)]^{1/2}$.

Table 2. Selected Bond Lengths and Bond Angles of $[\mathbf{2}](PF_6)$

Bond	Lengths/Å	Bond	Lengths/Å
Pt1-Cl1	2.270(2)	C5-O1	1.20(1)
Pt1-P1	2.255(2)	C6-O2	1.19(1)
Pt1-N1	2.079(6)	C5-C6	1.54(1)
Pt1-N2	2.043(6)		
Bond	Angles/deg	Bond	Angles/deg
Cl1-Pt1-P1	91.18(8)	P1-Pt1-N2	96.2(2)
C11-Pt1-N1	92.6(2)	N1-Pt1-N2	80.0(2)

chlorine and two nitrogen atoms coordinated on $[PtCl_2(Me_2-phenO_2-N,N')]$ largely deviate out of plane due to the steric repulsion induced by two Me-groups. ¹⁶ In addition, Me₂phen of $[PtI_2(Me_2phen)(PPh_3)]$ (Me₂phen = 2,9-dimethyl-1,10-phenanthlorine) is linked to Pt in a monodentate manner. ¹⁹ The square-planar structure of $[2]^+$ exhibits less steric repulsion between phenO₂ bonded to Pt with the diimine group and bulky PPh₃. The Pt–N distances (Pt1–N1, 2.079(6) Å and Pt1–N2, 2.043(6) Å) are close to those of previously reported crystal structures having a Pt–bipyridyl skeleton. ²⁰ The Pt–Cl bonding (Pt1–Cl1, 2.270(2) Å) is also in good agreement with those of [PtCl(bpy)(L)] and [PtCl(phen)(L)] (L = extra ligand)

complexes.²¹ The Pt–P distance (Pt1–P1, 2.255(2) Å) is comparable to that of [PtMe(Me₂phen)(PPh₃)]⁺.²² The two C=O double bond distances are 1.20(1) and 1.19(1) Å, which are in good agreement with that of metal–free phenO₂ (1.209(3) Å)^{9a} and those of other metal–phenO₂ complexes.^{13,23} Platinum(II)–polypyridyl complexes tend to form π - π or Pt- π Pt stacking columns in the crystal structure.²⁴ On the other hand, the Pt- π Pt atomic distance of 7.55 Å in the crystal structure of [2]⁺ is an implication of no interaction between the neighboring two Pt metals. In addition, PF₆⁻ is positioned between two phenO₂ ligands of the adjacent molecules. The bulky PPh₃ of [2]⁺ may be responsible for blocking the formation of intermolecular Pt- π Pt and π - π stackings in the solid state.

Electronic Spectra of Pt(II) Complexes in the Presence of Lewis Acids. The complex $[2]^+$ has an N,N'-chelate ring with a free quinoid group. We obtained [Pt(phenO₂-O,O')-(PPh₃)₂] (3), which has an O,O'-chelate ring with a free diimine site, by the reaction of [Pt(PPh₃)₄] with phenO₂ in CH₂Cl₂.²⁵ The difference in the coordination modes of phenO₂ in $[2]^+$ and 3 results from the strong π acceptor character of a quinoid group and strong σ donor ability of a diimine group. The complex $[2]^+$ bearing the $Pt^{II}(phenO_2-N,N')$ framework is formed by a nucleophilic attack of the diimine of phenO₂ on Pt(II). On the other hand, the formation of 3 with the Pt^{II}(phenO₂-O,O') framework is ascribed to an electrophilic attack of the quinone of phenO2 to Pt(0) and the subsequent intramolecular electron transfer from the low valent metal center to the ligand. The basicity of the quinoid moieties of phenO₂ in [2]⁺ would be substantially decreased compared with that of free phenO2, whereas the electron densities of the diimine group of phenO₂ in 3 would be greatly increased compared with that of free phenO₂. The basicity of the diimine and quinoid groups of [2]+ and 3 were estimated by the electronic absorption spectra in the presence of Lewis acids.

An addition of 1.0 equiv of aqueous HClO₄ solution to a CH₃CN solution of [2](SbF₆) did not cause any changes in the electronic absorption spectrum or ESI-MS spectrum of the complex. Therefore, the interaction between [2]⁺ and HClO₄ is extremely low or practically not detected under the experimental conditions. On the other hand, an addition of HClO₄ to a CH₃CN solution of 3 resulted in degradation of the complex, because the parent peak of the ESI-MS spectrum of 3 $(m/z 930 ([3]^+))$ decreased in intensity with increasing amounts of HClO₄ added and a new peak of m/z 796 corresponding to [PtCl(CH₃CN)(PPh₃)₂]⁺ appeared, even in the presence of 0.1 equiv of HClO₄ in CH₃CN. Taking into account that the phenO₂ of 3 is linked to Pt(II) with the catecholato form, dissociation of phenO₂ from Pt(II) in the presence of HClO₄ is explained by the protonation of the catecholato oxygens. Thus, HClO₄ is too strong a Lewis acid to evaluate the basicity of the diimine group of 3. Therefore, we examined the interaction of [2]⁺ and 3 with LiBF₄ in place of HClO₄ in CH₃CN. As in the case of the interaction between [2]⁺ with H⁺, the absorption spectra of [2](SbF₆) in the presence of a large excess of LiBF4 in CH3CN did not give any evidence of an interaction between [2]⁺ and Li⁺ in the solution. On the other hand, an addition of LiBF4 to a CH3CN solution of 3 resulted in a decrease in the absorption band at 373 nm, and an appearance of two new bands at 328 and 396 nm. The spec-

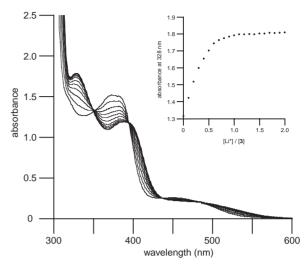


Fig. 3. The electronic absorption spectra of **3** (0.200 mM) in the presence of various amounts of LiBF₄ (0–1.0 equiv) in CH₃CN. Inset: titration of **3** with LiBF₄ in CH₃CN at 328 nm.

tral changes almost stopped after one equiv of Li⁺ was added to the solution, during which four isosbestic points appeared at 350, 394, 436, and 480 nm (Fig. 3). Further addition of Li⁺ to the solution did not cause any changes in the spectrum. Thus, Li⁺ and **3** almost quantitatively formed a 1:1 adduct in CH₃CN (Eq. 3).²⁶ An addition of HClO₄ to **3** caused degradation of

the complex, probably due to protonation of the catecholato oxygens. On the other hand, the most reasonable binding site of Li⁺, is the diimine nitrogen of **3**, since only a 1:1 adduct is formed between Li⁺ and **3**, even in the presence of a large excess of Li⁺ in solution. Indeed, the ESI-MS spectrum of **3** in the presence of 10 equiv of Li⁺ showed a peak of m/z 937 assignable to **3**·Li⁺ besides a parent peak (m/z 930 ([**3**]⁺)). The phenO₂ ligand bonded to Pt(II) with the two-electron reduced form (see below) endowed such a strong basicity to the diimine group to bind Li⁺ almost quantitatively.

We also examined the interaction between free phenO₂ and Li⁺ to evaluate the degree of the increase in basicity of the phenO₂ of 3 compared with that of the free ligand. The electronic absorption spectrum of phenO2 shows two absorption bands at 296 and 308 nm in CH₃CN (Fig. 4). An addition of LiBF₄ to the solution results in an increase in the intensity of those bands. The two bands continuously increase in intensity with increasing amounts of Li+ added to the solution and merges at 308 nm in the presence of more than 100 equiv of Li⁺. The appearance of an isosbestic point at 279 nm that remained unchanged irrespective of the amounts of LiBF4 is indication of the 1:1 adduct formation between phenO2 and Li^+ in CH₃CN (Eq. 4). The binding constant K for the adduct formation between phenO2 and Li+ was calculated according to Eqs. 4-6. In the above equation, A₀ and A are the absorbance of phenO₂ at 296 nm in the absence and in the presence

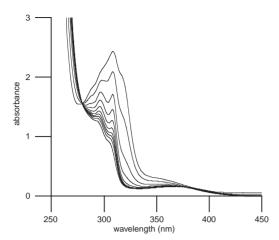


Fig. 4. The electronic absorption spectra of phenO₂ (0.305 mM) in the presence of various amounts of LiBF₄ (0-200 equiv) in CH₃CN.

$$phenO_2 + Li^+ \rightleftharpoons phenO_2 \cdot Li^+, \tag{4}$$

$$K = \frac{c_{\text{phenO}_2\text{Li}}}{c_{\text{phenO}_2} \cdot c_{\text{Li}}},\tag{5}$$

$$K = \frac{c_{\text{phenO}_2 \text{Li}}}{c_{\text{phenO}_2} \cdot c_{\text{Li}}},$$

$$\frac{1}{(A_0 - A)} = \frac{1}{K \cdot X} \cdot \frac{1}{c_{\text{Li}}} + \frac{1}{X},$$
(5)

where $X = c_{phenO_2} \cdot \mathcal{E}_{phenO_2} + c_{phenO_2Li} \cdot \mathcal{E}_{phenO_2Li}$

of Li⁺, respectively, c_{phenO_2} , c_{Li} , and $c_{\text{phenO}_2\text{Li}}$ are the concentrations of phenO₂, Li⁺, and complexed phenO₂•Li⁺, respectively, and \mathcal{E}_{phenO_2} and \mathcal{E}_{phenO_2Li} are the molar extinction coefficients of phenO₂ and complexed phenO₂•Li⁺, respectively. A plot of $1/(A_0 - A)$ against $1/c_{Li}$ gave a straight line in the presence of more than 20 equiv of Li^+ . The binding constant K obtained from the intercept was 3.6 M⁻¹. The binding constant for the adduct formation with Li⁺ therefore increases in the order of $[2]^+ \ll \text{phenO}_2 \ll 3$. The trend is apparently proportional to the electron density of the diimine moiety depending on the amount of electrons transfered from the o-quinoid part of phen O_2 .

Redox Behavior of [2]+ and 3 in the Absence and the **Presence of Li⁺.** The cyclic voltammogram (CV) of [2](SbF₆) in CH₃CN displayed two reversible redox couples at $E^{1}_{1/2}$ = $-0.02 \,\mathrm{V}$ and $E^2_{1/2} = -0.75 \,\mathrm{V}$ (vs SCE), and an irreversible cathodic wave at $E^3_{pc} = -1.80 \,\text{V}$ (Fig. 5a). The equilibrium electrode potential (E_{rest}) of the solution (+0.32 V) also supported the quinoid form of phenO₂ in [2]⁺. The reversible redox reactions at $E^{1}_{1/2}$ and $E^{2}_{1/2}$ are assigned to the ligandbased quinone/semiquinonato and semiquinonato/catecholato couples, respectively, and the irreversible one at E^{3}_{pc} is associated with the reduction of the central metal. Based on the redox potential of free phenO₂ ($E^{1}_{1/2} = -0.45 \text{ V}$ and $E^{2}_{1/2} = -1.25$ V),5a coordination of phenO2 on Pt(II) resulted in an anodic shift of the redox potential of the ligand by about 400-500 mV due to the electron donation from phenO2 to Pt(II) through the diimine chelate. As discussed above, the electronic absorption spectra of [2]⁺ did not display any interactions with Li⁺ in CH₃CN.²⁷ On the other hand, an addition of 1 equiv of LiBF₄ to the CH₃CN solution of [2]⁺ caused an anodic shift of the redox couple at $E^{1}_{1/2} = -0.02 \,\text{V}$ and $E^{2}_{1/2} = -0.75 \,\text{V}$ by 850 and 120 mV, respectively (Fig. 5b), and the two redox couples

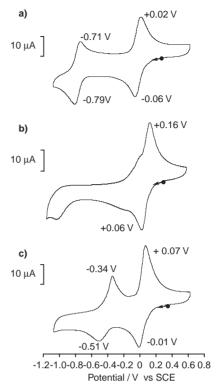


Fig. 5. Cyclic voltammograms of [2]⁺ (1.0 mM) in the absence (a) and the presence of 1.0 equiv of LiBF₄ (b) and NaClO₄ (c) in CH₃CN containing 0.1 M ⁿBu₄NPF₆. The dot in the voltammograms is the equilibrium electrode potential of the solutions.

fused into one peak at $E_{pa} = +0.16 \text{ V}$ and $E_{pc} = +0.06 \text{ V}$. Further addition of Li⁺ hardly had any influences on the pattern of the CV, indicating that the phen O_2 of $[2]^+$ has no ability to bind Li+. However, Li+ greatly facilitates the two-electron reduction of [2]⁺ through an ECE mechanism. As a result, the resultant $[2]^-$ quantitatively binds Li⁺ with the O,O'-chelate. As in the case of LiBF4, NaClO4 also caused anodic shifts of the redox potentials of [2]+, since two redox waves appeared at $E_{\rm pc} = -0.01 \,\mathrm{V}$ and at $E_{\rm pc} = -0.51 \,\mathrm{V}$ in the presence of 1 equiv of NaClO₄ in CH₃CN. However, the appearance of a spike anodic wave at $E_{pa} = +0.07 \text{ V}$ in the reverse potential scans is an implication of adsorption of the adduct between [2] and Na⁺ on the surface of the glassy carbon working electrode due to low solubility of the adduct (Fig. 5c).

The CV of 3 in CH₃CN displayed the reversible catecholato/semiquinonato redox couple and an irreversible semiquinonato/quinone one at $E^{1}_{1/2} = +0.28 \text{ V}$ and $E^{2}_{pa} = +0.81 \text{ V}$, respectively.6 The appearance of the equilibrium electrode potential of the solution ($E_{rest} = +0.03 \,\mathrm{V}$) at the negative site of the $E^{1}_{1/2}$ and E^{2}_{pa} redox reactions indicates that phenO₂ is linked to Pt with the catecholato form, not the semiquinonato one. The irreversible semiquinonato/quinone redox couple in the CV of 3 is ascribed to lability of the quinoid structure of 3 due to the extremely weak electron donor ability of the quinoid group.⁶ In contrast to [2]⁺, 3 appears to bind Li⁺ in CH₃CN quantitatively. The redox wave at $E^{1}_{1/2} = +0.28 \,\mathrm{V}$ gradually shifted to positive potentials with increasing the concentration of Li⁺ (Fig. 6). The shift of the redox waves almost stopped in

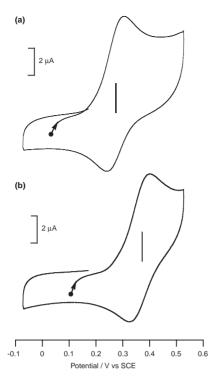


Fig. 6. Cyclic voltammograms of **3** (0.50 mM) in the absence (a) and the presence of 1.0 equiv of LiBF₄ (b) in CH₃CN containing 0.1 M n Bu₄NPF₆. The dot in the voltammograms is the equilibrium electrode potential of the solutions.

the presence of 1.0 equiv of LiBF₄, where the catecholato/ semiquinonato redox couple of 3 is observed at $E^{1'}_{1/2} =$ +0.37 V. The fact that 3 forms only the 1:1 adduct with Li⁺ indicates that Li⁺ does not attack the catecholato oxygens of 3. It is worthy to note that the difference in the redox potentials of the catecholato/semiquinonato couple of $[2]^+$ and 3 in the absence and in the presence of Li⁺ are $\Delta E^{1}_{1/2} = 850 \,\text{mV}$ and $\Delta E^{1}_{1/2} = 90 \,\text{mV}$, respectively. The change in the electron density of the quinoid moiety in [2]+ and 3 caused by the redox reactions at $E^{1}_{1/2}$ and $E^{2}_{1/2}$ is apparently much larger than that of the diimine one, because the phenanthroline moiety of phenO₂ plays a central role in the ligand-based redox reactions. As a result, the basicity of the quinoid oxygens exhibits great changes upon the phenO2 localized redox reaction, whereas the redox reactions have minor effects on the change of the electron density of the diimine group.

Experimental

Materials. Commercially available [K_2 PtCl₄] and triphenyl-phosphine were used without further purification. The tetrabutyl-ammonium hexafluorophosphate used as a supporting electrolyte for electrochemistry was recrystallized from hot ethanol. 1,10-Phenanthroline-5,6-dione (phenO₂),²⁸ [PtCl₂(PhCN)₂],²⁹ [PtCl₂-(phenO₂-N,N')] (1),¹⁵ and [Pt(phenO₂-O,O')(PPh₃)₂] (3)²⁵ were prepared according to the literature.

Preparations of Complexes. [PtCl₂(phenO₂-*N*,*N*')]•DMSO, [1]•DMSO: To a dimethyl sulfoxide solution (4 mL) of [PtCl₂-(PhCN)₂] (54.5 mg, 0.115 mmol) was added a dimethyl sulfoxide solution (2 mL) of 1,10-phenanthroline-5,6-dione (24.2 mg, 0.115 mmol). On standing overnight at room temperature, yellow crys-

tals formed out of the reaction mixture. The crystals were collected by filtration, washed with diethyl ether and dried in vacuo. Yield: 47.1 mg (0.085 mmol, 74%). Anal. Found: C, 30.07; H, 2.20; N, 4.99%. Calcd for $C_{14}H_{12}Cl_2N_2O_3PtS$: C, 30.33; H, 2.18; N, 5.05%. 1H NMR (500 MHz, DMSO- d_6 , rt, δ /ppm): 8.07 (dd, J(H-H) = 6.0, 8.0 Hz, 2H, phenO₂), 8.84 (dd, J(H-H) = 1.0, 8.0 Hz, 2H, phenO₂), 9.59 (dd, J(H-H) = 1.0, 6.0 Hz, 2H, phenO₂).

 $[(PtCl)(phenO_2-N,N')(PPh_3)](SbF_6), [2](SbF_6):$ A methanol suspension (80 mL) of [PtCl₂(phenO₂-N,N')] (82.3 mg, 0.173 mmol) and triphenylphosphine (45.3 mg, 0.173 mmol) was warmed to 50 °C for 1.5 h under N₂. The resulting yellow solution was filtered to remove unreacted [PtCl₂(phenO₂-N,N')]. The addition of a saturated aqueous solution of NaSbF₆ to the filtrate precipitated yellow solids, which were separated by filtration, washed with water and then diethyl ether, and dried in vacuo. Yield: 111.1 mg (0.118 mmol, 68%). Anal. Found: C, 37.78; H, 2.37; N, 2.94%. Calcd for C₃₀H₂₁ClF₆N₂O₂PPtSb • (H₂O): C, 37.66; H, 2.42; N, 2.93%. ¹H NMR (500 MHz, CD₃CN, rt, δ /ppm): 7.29 (m, 1H, phenO₂), 7.54 (m, 6H, PPh₃), 7.66 (m, 3H, PPh₃), 7.89 (m, 6H, PPh₃), 8.06 (m, 1H, phenO₂), 8.15 (m, 1H, phenO₂), 8.67 (d, $J(H-H) = 8.0 \,\text{Hz}$, 1H, phenO₂), 8.88 (d, $J(H-H) = 8.0 \,\text{Hz}$, 1H, phenO₂), 9.74 (m, 1H, phenO₂). IR (ν /cm⁻¹, KBr): 1709 (CO). ESI-MS $(m/z, \text{ CH}_3\text{CN})$: 703 $([2]^+)$. UV-vis $(\lambda_{\text{max}}/\text{nm} \ (\varepsilon/z)$ M^{-1} cm⁻¹), CH₃CN): 310 (9100).

[(PtCl)(phenO₂-N,N')(PPh₃)](BF₄), [2](BF₄): The complex was prepared by a similar method to that of [2](SbF₆) using NaBF₄ instead of NaSbF₆. Yield: 64%. Anal. Found: C, 45.11; H, 2.86; N, 3.48%. Calcd for $C_{30}H_{21}BClF_4N_2O_2PPt \cdot 0.5(H_2O)$: C, 45.11; H, 2.78; N, 3.51%.

[(PtCl)(phenO₂-N,N')(PPh₃)](PF₆), [2](PF₆): The complex was prepared by a similar method to that of [2](SbF₆) using NH₄PF₆ instead of NaSbF₆. Yield: 81%. Anal. Found: C, 42.33; H, 2.62; N, 3.46%. Calcd for C₃₀H₂₁ClF₆N₂O₂P₂Pt: C, 42.49; H, 2.50; N, 3.50%.

Measurements. ¹H NMR was performed with a JEOL GX-500 spectrometer. Infrared spectra were recorded on a Shimadzu FT IR-8100 spectrophotometer. ESI-MS spectra were measured with a Shimadzu LCMS-2010 liquid chromatograph mass spectrometer and a Waters Micromass LCT. Elemental analyses were carried out at the Research Center for Molecular-Scale Nanoscience, Institute for Molecular Science. Cyclic voltammetry was performed with an ALS/Chi model 660 electrochemical analyzer. Cyclic voltammograms were recorded at a scan rate of $100 \, \mathrm{mV \, s^{-1}}$ at room temperature using a glassy carbon working electrode, a Pt wire counter electrode and a Ag/AgNO₃ (0.01 mmol dm⁻³) reference electrode. All potentials were converted to SCE ($E_{\mathrm{SCE}} = E_{\mathrm{Ag/Ag^+}} + 0.327 \, \mathrm{V}$). UV–vis spectra were recorded on a Shimadzu UV-3100PC UV–vis–NIR scanning spectrophotometer and a Hewlett-Packard 8453 diode array spectrophotometer.

Crystallography. Pale yellow crystals of [2](PF₆) were obtained by vapor diffusion of diethyl ether into an acetonitrile solution of the complex. The crystal data of [2](PF₆) are summarized in Table 1. Data for [2](PF₆) were collected on a Rigaku/MSC Mercury CCD diffractometer using graphite-monochromated Mo K α radiation ($\lambda = 0.71070 \,\text{Å}$) at 173 K, and processed using CrystalClear.³⁰ A numerical absorption correction was carried out and the data were corrected for Lorentz and polarization effects. The structure was solved by direct methods (SIR92)³¹ and refined by full-matrix least-squares refinement on F^2 . All non-hydrogen atoms were refined anisotropically. All hydrogen atoms were located in the calculated positions and not refined. All calculations were performed using the teXsan crystallographic software

package.³² Crystallographic data have been deposited with Cambridge Crystallographic Data Centre: Deposition number CCDC-280916 for compound [2](PF₆). Copies of the data can be obtained free of charge via http://www.ccdc.cam.ac.uk/conts/retrieving. html (or from the Cambridge Crystallographic Data Centre, 12, Union Road, Cambridge, CB2 1EZ, UK; Fax: +44 1223 336033; e-mail: deposit@ccdc.cam.ac.uk).

Supporting Information

Tables of crystallographic data (Table S1), selected bond distances and angles (Table S2), and the ORTEP drawing (Fig. S1) for [1]•DMSO are formatted in PDF files. This material is available free of charge on the web at: http://www.csj.jp/journals/bcsj/.

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